Prospecting, Geochemical, Geophysical, Geological, Trenching and Diamond Drilling

of the

Ace, Frank Creek , SCR and Peripheral Properties

Little River Area,
Cariboo Mining Division, British Columbia
52° 45' N; 121° 15'W
NTS

Work was done on the following claims: Ace 15-19, 22-27, 33-44, Ace 57, 58, 60, 62, 64, 67-72, 74,76,78, 82-84, LED 14,16,18 Abracad 2, Bill 1, Boo 1 & 2, Jess 1-4, Big Gulp 1-3 & 5-8, Frank, F3, F4, K 13, K 7, Maud 5,7,8, VC 1,6,3,11, Mag 2,3,5, PG 1,2,4,9 & Rol 1-3

Owned by

BARKER MINERALS

Report prepared by Louis E. Doyle
from appended reports:
Perry (2002), Wild (2002) & Walcott (2002)

March 28, 2003

Barker Minerals Ltd.
March 2003
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1.0 Introduction

Work on the property to date includes geological mapping, stream-sediment and soil surveys, ground magnetometer and EM studies (VLF-EM, HLEM), airborne magnetometer and EM studies, prospecting, and, in some areas of potential economic interest, trenching and diamond drilling.

2002 exploration work consisted of prospecting (54 rock samples collected), soil sampling (63 soil samples collected, which have yet to be analyzed), geophysical surveys, geological mapping, trenching, and drilling which is detailed in appended Company reports: Trenching and Drilling Report (Wild 2002), Preliminary Geophysical report (Walcott 2002), and Report on Exploration (Perry 2002).

The Ace Prospect mineralization and Frank Creek and SCR mineral showings are of the VMS type. The Cariboo Prospect hosts a replacement style Zn-Pb deposit in carbonate rocks of the Cariboo terrane. Scattered stream-sediment anomalies situated within the claims have not been followed up in detail. The Quesnel Platinum Project areas may contain some of the source mafic and ultramafic rocks for PGE minerals occurring in numerous placer platinum occurrences associated with the Quesnel River and its tributaries. These claims also host mafic volcanic rocks that in one area contain anomalous concentrations of copper.

1.1 Property

The mineral exploration property consists of 4,092 mineral claim units, being approximately 260,000 acres or 105,222 hectares (Figure 1). The mineral claims comprising the property are owned by and registered in the name of Barker Minerals Ltd. or Louis Doyle in trust for Barker Minerals Ltd. (Perry 2002-Appendix 1). The property contains 16 mineral exploration project areas, some of which are currently under active exploration, including the Ace, Frank Creek, Sellers Creek Road (SCR) and Quesnel Platinum project areas. The mineral claim tenure numbers and expiry dates of the mineral claims comprising Barker Minerals' exploration property are listed in Perry 2002-Appendix 1.

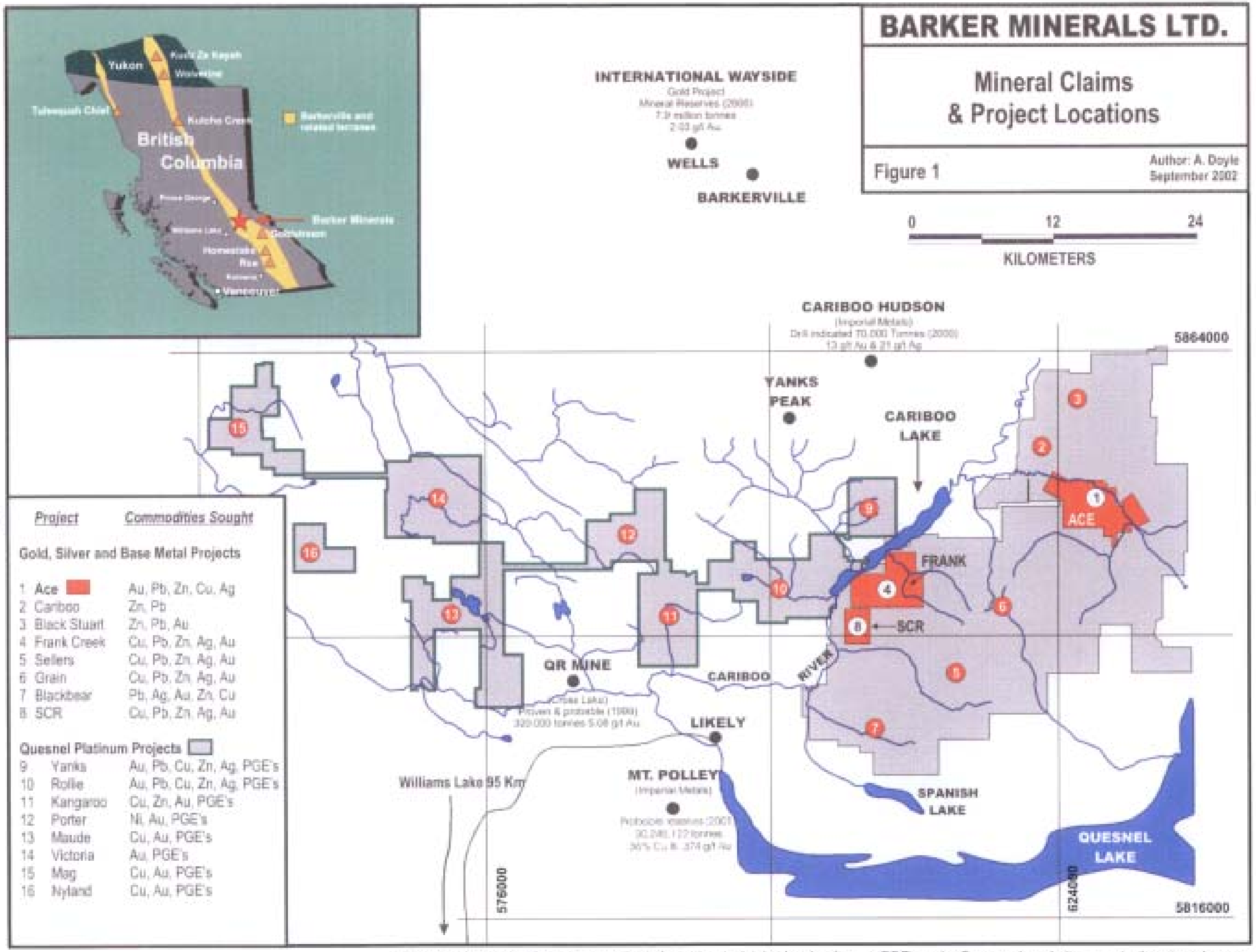
1.2 Location and Access

The center of the property is situated 95 km northeast of the city of Williams Lake, B.C., the nearest supply center, and 34 km northeast of Likely, B. C., the nearest settlement.

Exploration work conducted by Barker Minerals Ltd. to the present date has been conducted under the authority of Mineral Exploration and Reclamation Permit MX-10-155. The local BC Inspector of Mines has indicated to Barker his satisfaction with Barker's progressive reclamation of trench and drill sites, and indicated also his satisfaction with the amount of the current reclamation bond Barker has supplied, without need for any further increase in the bond in respect to additional work planned.

1.3 Regional Economic History

Gold was discovered in the Barkerville-Wells area in 1858. Historical production totaled 3.7 million troy ounces, as 1.9 million ounces from placers and 1.8 million ounces from 2.7 million short tons of underground ore. The historic Bullion Pit near Likely produced 175,700 ounces of gold from 200 million tons of gravel and about 1/100th as much platinum.



* Enlargeable detailed figures are available for review in Adobe Acrobat format (PDF) on the Company's website: www.barkerminerals.com

Much of the lode and placer gold production from the Wells and Barkerville areas occurred in the eastern part of the Barkerville terrane, where most of the important present day lode gold prospects also occur. At the historic mines, the strata trend 315°, dip 45°NE and are cut by north to northeast-trending normal faults dipping 60°E. The zones of economically important quartz veins are contained in graphite-bearing layers near a contact with carbonate-bearing layers. Two sets of quartz veins contain pyrite and gold: transverse veins striking 030° and diagonal veins striking 070°; these contained 15-25% pyrite and 0.3-0.5 oz/t Au. Two other sets of quartz veins do not contain pyrite or gold. Pyrite in altered wall rock contained proportionally less gold. Replacement bodies of auriferous pyrite in carbonate rocks (mainly in the Island Mountain Mine north of Jack of Clubs Lake) accounted for nearly 1/3 of the lode gold mined. Other common sulphides in the mines are arsenopyrite, galena, sphalerite, cosalite (PbBiS), scheelite and bismuthinite (BiS_2). Free gold commonly occurs with nests of cosalite. Gangue minerals are quartz and ankerite/siderite. Wallrock alteration minerals are ankerite and sericite.

1.3.1 Frank Creek Area

Frank Creek has hosted sporadic placer mining since the turn of the century, the latest periods being 1984 to 1986 and 2000 to 2001. During early placer mining, several massive sulphide boulders containing up to 9.3% Zn+Pb (Formosa 1989) were found in a thin, clay-rich layer situated near the base of the alluvial material where Frank Creek canyon opens onto a delta.

During 1988, Formosa Resources Corp. and Rio Algom Mines conducted grid-controlled soil sampling, VLF-EM geophysical surveys and limited geological mapping. As a result of this work, they identified several coincident geophysical and soil geochemical anomalies. A trench excavated on a coincident Cu-Zn-Ag soil anomaly encountered rusty soil and lenses of ferricrete containing pyrite-rich fragments that contain also minor chalcopyrite, galena and sphalerite. On the slope above the above-mentioned alluvial massive sulphide boulders, 22.4 km of grid lines were cut and soil sampling was conducted at 25-m intervals, yielding some anomalous Cu, Pb, Zn and Ag concentrations but no outcropping bedrock mineralization.

During 1991, helicopter-facilitated magnetic, electromagnetic (EM) and radiometric surveys by Rio Algom identified seven areas containing anomalous EM signatures in the Frank Creek area, F-1 to F-7 target areas (Figures 5a and 5b).

During 1996, two, vertical, percussion holes were drilled along the D-Road switchback near the center of the F-1 target area by R. Yorston, a previous operator. They yielded poor recovery, mainly of very fine particles and minor amounts of rock chips of black and green phyllite. Both holes returned anomalous Cu, Pb, Zn, Ag and Au concentrations in the upper 18.3 m, with the best interval being 3.04 m averaging 1770 ppm Cu, 750 ppm Pb, 2970 ppm Zn, 4.8 ppm Ag and 790 ppb Au.

During the period covered by this report, the Company conducted geological and geophysical exploration work, including trenching, rock sampling, localized detailed mapping at the Frank Creek VMS discovery outcrop, diamond drilling and assaying (items 11, 12, 13 Perry-2002 report).

1.3.2 Cariboo Zn-Pb Deposit

Barker Minerals' Cariboo Prospect, also known as the Maybe Prospect, was explored during 1987 by Gibraltar Mines Ltd. The prospect contains three main stratiform lenses of ankerite, quartz, sphalerite, galena and minor pyrite enclosed in limestone-rich strata of probable Middle Devonian age. The large Zn/Pb ratio, moderate silver content and low gold content are similar to those of many carbonate-hosted, replacement Zn-Pb deposits, including those in the Early Cambrian platform carbonates of the Kootenay arc. Limited bulldozer trenching and 21 diamond drill holes by Gibraltar Mines resulted in an in-house mineral inventory estimate of 400,000 tonnes at a grade of 4% Zn+Pb. This is not an appropriate category and does not comply with the requirements of current securities legislation. It has been recommended that limited confirmatory fieldwork and limited confirmatory drilling be done in order to define and categorize the known mineralization to current day industry and regulatory standards.

1.3.3 Quesnel Platinum Project Areas

Some placer deposits associated with the Quesnel River and some of its tributaries emanating from the north and east contain potentially significant concentrations of platinum group metals (Rublee, 1986 and Figure 2, Perry-2002). The highest concentration was obtained from a pan concentrate sample collected from Twenty-Mile Creek that assayed 2195 g/t Pt, 2210 g/t Pd and 1440 g/t Os. In this concentrate, the platinum group minerals (PGMs) were found as minute metallic grains within larger grains of magnetite and chromite.

A regional stream survey conducted by the BC Geological Survey during 1980 (GSC Open File 776), with results re-analyzed during 1997 by neutron activation (BC Geological Survey; BC RGS 50), showed anomalous concentrations of Cu, Ni, Cr, Co and Au in and around areas of ultramafic rocks that are situated partly along the Eureka fault and partly within the Quesnel terrane. An airborne magnetometer survey indicated several strong anomalies associated with ultramafic rocks located along the Eureka fault and at known locations of magnetite-bearing mafic and ultramafic rocks in the Quesnel terrane (Dome Exploration, 1981). During 1984, helicopter-facilitated multi-frequency EM, VLF-EM and magnetometer surveys were conducted in the Maud Lake, Victoria Creek and Trumph areas and many EM and magnetic anomalies were detected and recommended for further geological and geochemical investigation (BC Assessment Report #12780). During the late 1980s, QPX Minerals drilled two holes into ultramafic bodies at its Maude Lake property. Both holes, a few hundred metres apart, contained geochemically anomalous concentrations of Pt up to 35 ppb (1988, BC Assessment Report #17598). At the bottom of one of the drill holes was a 10-cm intersection of chalcopyrite-pyrite massive sulphide for which no assays are available. Mandella Resources conducted ground geophysical and geochemical soil surveys west of this area in the mid-1980's and identified VLF-EM geophysical anomalies and Pb, Zn, Cu and Au soil geochemical anomalies, which in some instances were coincident (BC Assessment Report #14816). Some of the PGM contents of the above mentioned Quesnel River placer deposits are probably derived from ultramafic rocks situated along the Eureka fault between the Quesnel terrane and the Barkerville terrane and from mafic and ultramafic rocks in the Quesnel terrane. Barker Minerals' claims in this area were staked because of potentially favorable geology, the recent dramatic increase in the price of PGMs and the apparent lack of previous PGE exploration in this potentially favorable area.

Reconnaissance geological, geochemical and airborne geophysical surveys were conducted over portions of the area by several companies in the 1980s, mainly during exploration for porphyry copper deposits. Some of these studies identified Cu-Pb-Zn-Ag soil anomalies that are coincident with HLEM-indicated conductors, which have not yet been tested by trenching or drilling.

1.4 Geography and Physiography

Williams Lake is an intermediate-sized city, which is served by Highway 97, the B.C. Railway, a major hydroelectric power grid and a modern airport. By road, Likely is 65 km northeast of 150 Mile House on Highway 97. Access to the Ace, Frank Creek and SCR exploration areas is *via* gravel logging roads bearing northeast from Likely. The distances from Likely to the main showings are as follows: Ace, 45 km; Frank Creek, 25 km; SCR, 22 km. Driving time to the Ace prospect from Likely is forty-five minutes. Access to the Quesnel Platinum project is mainly *via* gravel logging roads southeast from Quesnel, a distance of 25 km. In Likely, Barker Minerals maintains a property that includes a house, a bunkhouse, a workshop and a few tents. The house serves as a field office.

The property is situated in the central part of the Quesnel Highland between the eastern edge of the Interior Plateau and the western foothills of the Columbia Mountains. This area contains rounded mountains that are transitional between the rolling plateaus to the west and the rugged Cariboo Mountains to the east. Pleistocene and Recent ice sheets flowed away from the high mountains to the east over these plateaus and down to the southwest (Cariboo River), west (Little River) and northeast (Quesnel Lake), carving U-shaped valleys. The elevation ranges from 700-1650 m.

Precipitation in the region is heavy, as rain in the summer and snow in the winter. Drainage is to the west *via* the Cariboo, Little and Quesnel Rivers to the Fraser River. Quesnel Lake, the main scenic and topographic feature in the region, is a deep, long, forked, glacier-carved lake with an outlet at 725 m elevation. Vegetation is old-growth spruce, fir, pine, hemlock and cedar forest in all but the alpine regions of the higher mountains (mainly above 1400 m elevation). Weldwood has been actively logging fir, spruce and pine in the area (e.g., logging clear-cuts shown in aerial photo Figure 5c, Perry 2002), principally during winters, and has provided outlines of existing and planned roads and cut-blocks in and near the project areas.

1.5 Previous Work and Summary

The property is located 95 km northeast of Williams Lake in Central British Columbia. The property contains the idle Providence Mine, classified as a 'Past Producer' (BC MinFile 093A 003) of silver, lead, zinc and gold. The property also contains the Cariboo (a.k.a. Maybe) Prospect, which is classified as a 'Developed Prospect' (BC MinFile 093A 110). This is reported to be a lead-zinc (Zn-Pb) replacement-style deposit estimated to contain approximately 400,000 tonnes at an estimated grade of 4% Zn+Pb, using a 1% Zn+Pb cutoff. The property contains the Ace VMS Prospect, which was discovered during 1993 by Louis Doyle, President and CEO of Barker Minerals and is host to what has been described by BC Geological Survey geologists as Besshi-type volcanogenic massive sulphide (VMS) mineralization and auriferous (gold-bearing) quartz veins (BC MinFile 093A 142). The property contains several known bedrock mineral occurrences, classified as 'showings' by the BC Geological Survey, including the Frank Creek VMS showing (BC MinFile 093A 152), the Sellers VMS showing (BC MinFile 093A 131), the Big silver-lead-gold showing (BC MinFile 093A 151), the Comin Throu Bear lead-zinc-silver showing (BC MinFile 093A 148), the Peacock gold-bearing quartz vein showing (BC MinFile 093A 133), the Maud alkalic porphyry copper-gold showing (BC MinFile 093A 119) and the Trump silver-lead showing (BC MinFile 093A 154). Both the Frank Creek VMS showing and Sellers Creek Road (SCR) VMS showing were discovered by Louis Doyle, President of Barker Minerals Ltd.

The eastern half of the property contains three VMS exploration project areas, the Ace, Frank Creek and SCR areas, each of which contain multiple exploration targets as indicated by geochemical, geophysical and geological data and which have been the focuses of Barker Minerals' exploration programs during the period 1993 to the present. The western half of the property contains the mineral claims hosting Barker Minerals' Quesnel Platinum Project.

Within the Ace project area surface geological, geochemical and geophysical surveys and two episodes of drilling [7 DDH (1260 m) in1998; 5 DDH (645 m) in 2002] have defined a belt of metamorphosed and deformed, felsic volcanic rocks containing massive and stringer sulphide mineralization, within which are anomalous concentrations of gold, silver, copper, lead and zinc. The belt is open along strike in both directions. Geophysical surveys have defined another major target located to the southeast of this belt in an apparently outcropless area containing encouraging soil geochemistry. Further exploratory trenching and drilling are recommended on these targets.

Within the Frank Creek project area a massive sulphide layer 1.2 m thick outcrops in a zone of overturned volcanic and sedimentary rocks. During 2002 a series of trenches were excavated in order to test several targets in and adjacent to the F-1 Target Area at the Frank Creek VMS project. Five exploratory trenches were excavated over a total distance of 289 metres up to 70 metres southeast of the discovery outcrop. Trench TR-BW-05 exposed the dark quartz eye phyllite host unit within which were found several small massive sulphide lenses within the projected strike extension of the mineralized zone exposed at the discovery outcrop. TR-BW-04 exposed copper stringer mineralization 60 metres southwest of the F-1 occurrence. Along with the mineralized exposure in TR-BW-10 excavated some 375 metres to the NW of the discovery outcrop, this exposure extends the known strike length of the mineralized zone to approximately 425 metres. The zone is open to potential extensions in both directions to the north and south and at depth. According to Wild (2002), former Chief Geologist of Goldstream Mine, near Revelstoke, B. C., the geological setting, mineralization and host rocks are all remarkably similar to the Goldstream Mine mineral deposit, which produced more than 2 million tonnes of ore at a grade in excess of 4.0% copper and 2.2% zinc.

Drill core from the initial exploratory drilling program [6 ddh (813 m)] at the Frank Creek project area contains intervals of Cu-Zn-Pb (+/- Au, Ag) massive sulphide mineralization that are significant examples of ore formation processes having occurred on the property. The mineralizations encountered in the drill core are similar to that exposed at the discovery outcrop where the discovery outcrop massive sulphide layer has been further exposed (3.5m in length and 1.5 metre wide) by trenching, and the local area mapped in detail (Wild, 2002). The concentrations of metals from grab samples of the outcrop massive sulphides ranged up to 4.4% Cu, 8.2% Zn, 1.1% Pb, 14.8oz/t Ag and 854 ppb Au. A chip sample across 5.77 metres exposed width assayed 0.82% Cu, 0.25 % Zn, 0.21% Pb and 44.3 g/t Ag including a .77 metre exposed width which assayed 2.1% Cu, .34% Zn, .11% Pb and 69 ppm Ag.

This Besshi-type VMS polymetallic mineralization also occurs in drill core in significant intervals (up to 0.4 metres) and contains significant concentrations of zinc, copper, silver, lead and gold (up to 3.4% Zn, 2.1% Cu, 2.8 oz/t Ag, 0.53% Pb and 746 ppb Au) within larger weakly mineralized units up to 52 metres wide in drill core intercepts that contain widespread disseminations of these metals. Several significant, combined geophysical and geochemical anomalies are present, one of which is situated within the area in which the massive sulphide outcrop occurs. Further exploratory trenching and drilling are recommended in order to test these specific targets for economic mineralization and in order to further define the extent of the mineralized zone.

The SCR project area contains semi-massive sulphide mineralization in altered volcanic rocks. This project area also contains coincident base-metal soil anomalies and HLEM geophysical anomalies in an area of sparse outcrop. A Maxmin geophysical survey comprised of 4.2 line kilometers was completed during the 2002 field season. In areas of geophysical and geochemical anomalies, prospecting was successful in discovering float boulders which assayed as high as 17.3% Zn and 6.4% Pb. Further surface exploration including trenching and bedrock sampling in this area is recommended, to be followed by initial exploratory drilling.

The Cariboo Zn-Pb deposits reported to be comprised of replacement style Zn-Pb mineralization hosted in carbonate rocks of the lower strata of the Cariboo terrane. Diamond drilling conducted during the1980's outlined a 400,000 tonne deposit grading 4.0% Zn+Pb (BC MinFile 093A 110). Further surface mapping should be conducted in this area in order to help gain an understanding of the deformation history of the deposit and the potential for extensions of the known zone. Compilation of all relevant data and limited diamond drilling is recommended in order to confirm the previous operator's drilling and in order to further define and investigate the size and economic potential of this deposit, which is open in both directions along strike and at depth.

The western part of the property (Quesnel Platinum Project area) was staked for its platinum group element (PGE) potential. It contains zones of anomalous and intense copper concentrations in mafic volcanic rocks and may contain some of the mafic to ultramafic source rocks for some of the platinum group minerals (PGMs) recovered from the predominantly gold-bearing placers associated with the Quesnel River and its tributaries. Further geochemical and geological surveys are recommended in order to explore for bedrock PGE mineralization and in order to explore the zone of mafic volcanic rocks containing anomalous concentrations of copper.

Exploration work conducted by Barker Minerals Ltd. to the present date has been conducted under the authority of Mineral Exploration and Reclamation Permit MX-10-155, which may be modified by amendments in order to facilitate future work such as that recommended within this report.

The estimated costs of the Phase 1 and Phase 2 programs are $1,780,000 and $4,000,000 respectively (Perry 2002).

2.0 Regional Geology

The regional geology was described by L.C. Struik (1988) and has been updated by F. Ferri, (2001). The Barkerville terrane is considered to be the northwest extension of the Kootenay terrane, which to the southeast overlies the Monashee metamorphic core complex, a large uplifted mass of high-grade paragneiss, quartzite and marble. The properties are on the flank of the northern, unexposed portion of this core complex. Northwest from the North Arm of Quesnel Lake the characteristic metamorphic minerals change from sillimanite through staurolite-kyanite, almandine garnet and biotite to chlorite northwest of the Ace claims. The garnet isograd runs northerly across the east-central part of the Ace group, while that of biotite is 30 km further northwest. Historic mines near Wells and Barkerville are in rocks of the greenschist facies. The age of both deformation and metamorphism is regarded as Mid-

Jurassic, which is interpreted as the time of collision of the North American plate to the east with a group of island arcs to the west. In the Little River area, four geological terranes are represented, most of which are dominated by marine sedimentary or metasedimentary rocks.

2.1 Barkerville Terrane

Most of the property area is underlain by marine strata of the Barkerville terrane, whose age is classified broadly as Late Proterozoic to Mid-Paleozoic. It is categorized by the Geological Survey of Canada as a subdivision of the Kootenay terrane. The region was deformed by intense, complex, in part isoclinal folding and overturning that produced an intimate interlensing of impure quartzite, siltstone, ankeritic dolomite, pelite and amphibolite. These rocks are cut by dikes and sills of metamorphosed diorite. Locally, stronger shear deformation produced mylonitic textures.

The northeastern third of this terrane is the main zone of economic interest in the Cariboo district. Struik described it as "gold-enriched", because it contains the historic Wells and Barkerville mines and the Cariboo Hudson deposit, 39 km and 18 km northwest of the Ace project area, respectively. This zone contains olive and grey micaceous quartzite and phyllite, amphibolite, marble, meta-tuff and meta-diorite sheets or sills. These descriptions are compatible with the rock types on the Ace project area, although the latter contains more metamorphosed felsic/intermediate volcanic rocks. Stratigraphic tops are unknown. The Barkerville terrane is cut by the Mid-Devonian Quesnel Lake gneiss (350 Ma), a coarse grained, leucocratic, biotite granitic gneiss with megacrysts of potassium feldspar. The main body of gneiss is 30 km long by 3 km wide and is elongated parallel to the eastern border of the Intermontane belt. Its contacts are in part concordant with, and in part perpendicular to, metamorphic layering. The Barkerville terrane hosts folded, sill-like masses up to 300 m thick of gneissic meta-diorite (400 Ma) and contains post-metamorphic anatectic pegmatite (86 Ma), particularly in a high-grade metamorphic aureole northwest of the North Arm of Quesnel Lake.

2.2 Cariboo Terrane

The northeastern part of the Little River area is underlain by marine peri-cratonic sedimentary strata of the Cariboo terrane. The Cariboo terrane consists mainly of limestone and dolomite with lesser siliceous, clastic, sedimentary rocks and argillite. Some geologists believe that the Cariboo terrane is a shallow, near-shore facies and the Barkerville is a deeper, offshore facies of the same erosion-deposition system. No rifting is suspected between the Cariboo terrane and the North American continent, in contrast to that between the Barkerville terrane and the North American continent.

The Cariboo and Barkerville terranes are separated by the regional Pleasant Valley thrust fault, which dips northeast moderately to steeply. It is reported by Struik (1988) to have moved the Cariboo block from the east over the Barkerville block along a strike length of over 100 km. In the map area, the fault cannot be found, suggesting that much of the movement attributed to it may have occurred by shearing in a broad zone along the "contact" between the two terranes.

Some of the carbonate layers in the lowest part of the Cariboo terrane (or upper part of the Barkerville terrane) are enriched in zinc and lead. Since the 1970's, preliminary exploration on stratiform Zn-Pb targets has been conducted in this area over a strike length of 23 km from the vicinity of the head of the North Arm, *via* Maeford Lake to the Cariboo (Maybe) prospect.

The Cariboo terrane was cut by the Jurassic-Cretaceous Little River stock, a medium-grained granodiorite grading to quartz monzonite. A normal fault along its southwest side (Little River fault) dips east and extends southeasterly to Limestone Point, on the western side of the North Arm of Quesnel Lake. It intersects, and in some literature has been confused with, the Pleasant Valley thrust. It moved chlorite-biotite metamorphic grade strata of the Cariboo terrane eastward to rest against staurolite-kyanite metamorphic grade strata of the Barkerville terrane.

2.3 Quesnel Terrane

A small southwestern portion of the Little River area is underlain by the Late Triassic to Early Jurassic, allochthonous Quesnel terrane. It was accreted to the North American continent, in part by subduction and in part by obduction. The Eureka thrust fault marks the boundary between the Quesnel and Barkerville terranes as well as that between the Intermontane and Omineca physiographic belts. The terrane is partly submarine and partly subaerial, consisting of volcanic and volcaniclastic rocks and co-magmatic intrusions, with minor carbonate lenses and related sedimentary rocks. Regionally, it hosts many important mineral deposits, mainly of Cu and Cu-Au, such as Highland Valley, Craigmont, Copper Mountain, QR and Mt. Polley. The Bullion Pit, from which 175,700 oz. of placer gold were produced, is near Likely just on the west side of the boundary between the Barkerville and Quesnel terranes.

2.4 Slide Mountain Terrane

Rocks of the allochthonous, Devonian to Late Triassic, Slide Mountain terrane underlie a very small part of the Little River area. Portions of these rocks were obducted, while others were subducted during collision of an oceanic plate with the continent. It is exposed east of Wells and Barkerville as the upper plate overlying the generally low-angle Pundata thrust fault. This fault it is nearly vertical where it crosses the southwestern part of the Little River area. Small slices of mainly mafic volcanic rocks and alpine-type ultramafic rocks of the Slide Mountain terrane occur in and parallel to the Eureka thrust. Minor lithologies include chert, meta-siltstone and argillite.

3.0 Drilling

The Company has conducted two episodes of drilling at its Ace VMS and vein gold project (Payne, 1998 and Wild, 2002) and one initial exploratory drilling program at its Frank Creek VMS project (Wild, 2002)

3.1 Ace VMS and Vein-gold Project

Barker Minerals' initial diamond drilling program at the Ace project (1998: 7 holes, ddh 98-01 to ddh 98-07, totaling 1260 m) targeted felsic strata and geophysical anomalies (chargeability highs, resistivity lows and magnetic highs) located within a zone suspected to be underlain partly by felsic rocks having exploration potential for massive sulphide deposits (Payne, 1999; BC Assessment Report). The holes were drilled between the main trench area and Colleen Road (Figures 4a, 4a-1, 4a-2 Perry, 2002). Summary drilling statistics are shown in Table 5a and Summary Drill Logs are shown in Table 5b of Perry 2002 report. Most of the holes were drilled to the south, dipping at -45°, which is approximately perpendicular to the regional trend of the metamorphic foliation. Two holes were drilled more steeply at -60°and -70° in order to attempt to penetrate specific geophysical anomalies. In some instances, it appears that the holes were not long enough to have reached the intended geophysical targets.

All drill holes of the 1998 program, except 98-05, intersected felsite, whose thickness ranged from 3.5 to 81.5 m. As the holes were drilled perpendicular or nearly so to the regional foliation, all thickness intervals are close to true thickness relative to the metamorphic event (Payne, 1999; BC Assessment Report). Because of the complex folding in the region, it is impossible to estimate true, pre-metamorphic thicknesses. Significant intersections and analytical results are presented in Table 5c of the 2002 Perry report. At the top of the main felsic section in DDH 98-03, a mineralized zone was intersected for 0.75 m, containing two intersections of massive sulphide mineralization, 0.25 m and 0.20 m long, separated by an intersection of calcite (probably limestone) 0.30 m thick. The massive sulphide mineralization carries apparently anomalous concentrations of Au, Ag, Cu, Zn, As, Se, Te, Sb, Bi, Mo and Cd, as determined by non-statistical consideration of the analytical data. The footwall alteration zone below the massive sulphide is anomalous in many of the same elements, generally with smaller concentrations than in the massive sulphide. As well, the footwall zone carries anomalous concentrations of Mo. Anomalous metal concentrations in the footwall decrease moderately to rapidly away from the top of the section of felsic rocks. Deeper in the footwall, many samples contain anomalous concentrations of Ag, Mo and Zn. In the centre of the felsite section, hole 98-07 intersected semi-massive sulphide mineralization for 0.36 m, containing anomalous concentrations of Ag, Cu, Bi and Se. Above and below this, the rocks contain anomalous concentrations of Ag, Zn, Mo, Bi and Se. Concentrations larger than 100 ppb Au, 500 ppb Ag, 200 ppm Cu, 70 ppm Pb and 150 ppm Zn are considered geochemically anomalous in this report.

During 2002, two diamond drill holes, ACE-02-01 and 02, were collared in the 16S Zone The first hole, Ace-02-01, tested a coincident max-min conductor and modest gravity anomaly and is located slightly up-ice from the high-grade (16.4% Zn/Pb) boulders found in previous programs. The top 49.0 metres consist of very siliceous quartz-muscovite schist and argillite. The next 10 metres consist of strongly calcareous marble, calc-schist, and argillite, immediately above a 6.5 metre section of "felsite". The felsite consists of plagioclase or albite with minor micas and quartz. The felsite in this area has concentrations in drill core up to 339 ppm Cu, 568 ppm Pb, and 156 ppm Zn. A narrow intercept (.07 metres) of Au/Bi/Te/W quartz vein assayed 745 ppb gold. Follow up geophysical surveys indicate that the possibility exists that this zone was faulted off where drilling occurred; further interpretation is required to determine the next program for this zone (Wild, 2002).

The second hole, Ace-02-02, tested a magnetic high anomaly but failed to explain the source of this anomaly. This strong, large magnetic anomaly is located up-ice from a zone of Au/Bi/Te/W quartz sulphide boulders that have gold concentrations up to 11 grams per tonne (g/t). It is anticipated that the next phase of drilling will test this high priority target for its economic gold potential. A 7.5 centimetre wide quartz vein at 159.6 metres near the end of hole Ace-02-02 assayed 692 ppb gold (Wild, 2002).

The other three drill holes in the Ace Project area, Ace-02-03 to 05, tested the 5N Zone (Figure 4a-2). All three intersected at least 40 metres of "felsite", which hosts VMS mineralization elsewhere on the property. The fine-grained nature of the rock, its intimate relationship to a series of thin marble bands, and almost regional extent suggest that it may be an exhalative horizon and an excellent target horizon for VMS deposits (Wild, 2002).

Two of the recent holes intersected significant sulphide mineralization near the top of the interval. Ace-02-03 intercepted 3.3 metres of semi-massive to massive sulphide mineralization with anomalous Cu/Pb/Zn metals. Mineralized felsite extended an additional 69 metres down hole below the strongly mineralized layer.

In drill hole Ace-02-04, a ten-metre interval of mineralized felsite between 43.8 - 53.8 metres, was highly anomalous for base and precious metal with concentrations up to 663 ppm Cu, 855 ppm Zn, 704 ppm Pb and 575 ppb Au. This hole was collared within 400 metres of a cluster of previously identified high grade quartz vein float boulders assaying up to 29 g/p/t Au, and zinc-lead mineralization (10% Zn, 2%Pb) in bedrock.

The last hole, Ace-02-05, was collared on L5N at 6+25W and drilled vertically to test a coincident ground magnetic high and subtle gravity anomaly. It was collared in the felsite unit, likely below the sulphide horizon (Wild, 2002).

Felsic volcanic rocks encountered during this drilling program consist mainly of extremely fine-grained plagioclase with minor biotite and/or muscovite. Many contain minor to abundant replacement and recrystallized patches of coarser grained plagioclase, with or without minor to abundant quartz, ankerite, muscovite and pyrite. Some surface samples, previously described as metamorphosed diorite, were reinterpreted as recrystallized and replaced felsic volcanic rocks. Some felsic volcanic rocks contain zones up to several metres wide of weak to strong biotitic alteration, some of which occur in broad, diffuse envelopes about quartz-sulphide veins. In these zones, pyrrhotite is replaced by coarser-grained porphyroblasts and lenses of pyrite.

Veins containing anomalous metal concentrations are of three types. Quartz-pyrrhotite veins in DDH 98-01 carry anomalous concentrations of Cu, Ag and Bi. Quartz-pyrrhotite-tourmaline veins in DDH 98-02 carry anomalous concentrations of Au, Ag, Cu, Zn, Bi, Se and Te. A quartz-pyrrhotite vein in DDH 98-07 carries anomalous concentrations of Au, Ag, Cu, Pb, Bi and Se. Quartz (+pyrrhotite) veins intersected in DDH02-01, DDH 02-02,DDH 02-03 and DDH 02-04 carry anomalous metal concentrations of Au, Ag, As, Sb, Bi, Te, W, Mo and Tl with concentrations of gold up to 764 ppb over 7 cm in DDH 02-01, 692 ppb over 7.5 cm in DDH 02-02 and 526 ppb over 25 cm in DDH 02-04. All six samples from DDH02-01 had geochemical patterns anomalous in Au, Ag, As, Sb, Bi, Te, W, Mo and Tl. A zone from 42.5 metres to 58.6 metres in DDH02-04 was anomalous in Au, Ag, As, Sb, Bi, Te, W, Mo and Tl. Numerous barren quartz and barren quartz-chlorite veins are present in drill core in many rock types and are particularly abundant in some intervals of quartzite and quartz-rich schist.

Comparison of drill-core data with previous analyses of boulders indicates that the massive sulphides and auriferous quartz-sulphide veins in the boulders and corresponding materials in the drill cores may have come from the same environment. This and the general angular appearance of the boulders and other large blocks of bedrock with which they occur in the overburden suggest strongly that the boulders are from a nearby bedrock source, probably slightly up-ice from their present locations.

3.2 Frank Creek VMS Project

During 2002, the Company conducted an initial exploratory diamond drilling program (7 ddh, 813m in total) in the vicinity of the Frank Creek massive sulphide discovery outcrop and nearby geological and geophysical targets (Table 6a;Table 6b 2002 Perry report). All of the initial drill holes were located within the F-1 Target Area (Figures 5a, 5b and 8a Perry 2002).

The drill core obtained during this initial exploratory program contains intervals of Cu-Zn-Pb (+/- Au, Ag) massive sulphide mineralization that are significant examples of ore formation processes having occurred on the property.

The mineralizations encountered in the core (Table 6c Perry 2002) are similar to that exposed at the discovery outcrop where a massive sulphide layer is exposed for some 3.5m length and 1.5 metre width (Lane 2000; Hunter Dickinson, 2000; and Hudson Bay Exploration & Development 2001). There, concentrations of metals from grab samples of the outcrop massive sulphides ranged up to 4.4% Cu, 8.2% Zn, 1.1% Pb, 14.8oz/t Ag and 854 ppb Au. A chip bedrock sample across .77 metres exposed width assayed 2.1% Cu, .34% Zn, .11% Pb and 69 ppm Ag.

This Besshi-type VMS polymetallic mineralization occurs in drill core in significant intervals (up to 0.4 metres) and contains significant concentrations (up to 3.4% Zn, 2.1% Cu, 2.8 oz/t Ag, 0.53% Pb and 746 ppb Au) within larger weakly mineralized units up to 52 metres wide in drill core intercepts that contain widespread disseminations of these metals.

Four of the 6 holes tested the Frank Creek mineralized horizon; the other two tested the immediate structural footwall, to a wide, conductive graphitic fault zone. FC-02-01, 05, and 06 intersected disseminated to semi-massive sulphides, dominantly pyritic, near the projected location of the mineralized horizon, downslope (northwest) of the original Frank Creek (F-1) Showing. FC-02-03 tested under the showing and intersected approximately 50cm of pyritic massive sulphide similar to that in the showing, caught in a large gougy fault zone. FC-02-02 and 04 tested strong max-min conductors and intersected significant graphitic shear zones in carbonate-altered metasedimentary rocks.

4.0 Interpretations and Conclusions

The Barker Minerals VMS and Vein Gold exploration Projects have been visited and examined, in various degrees of detail, by numerous independent professional geo-scientists and engineers, some associated with major mining companies or the BC Geological Survey. They have relayed to Barker their views on the exploration potential of the property and Barker Minerals' management and performance. The following are verbatim comments that Dr. Perry includes in his 2002 report in order to more fully appraise the reader of other professionals' views on Barker's property and Projects.

Charles R. Lammle, P. Eng. (1995):
"Realizing that he had made a good discovery, and that he was dealing with mineralization of complex mineralogy, he opened a line of communication for advice and support regarding the Ace property with senior provincial and federal government geologists. These geologists were solidly supportive, and results from the property continued to be encouraging."

Strathcona Mineral Services Ltd. (1998):
"Barker Minerals has assembled an impressive amount of information on the Ace and surrounding claims"

V. Preto, Ph. D., P. Eng., BC Geological Survey (1995):
"Congratulations on the report and all the work that has been done - both excellent"

T. Hoy and V. Preto, Geologists, BC Geological Survey (1996):
"Three deposit types were observed on the tour: Au-quartz veins, carbonite-hosted Zn-Pb and besshi-style massive sulphide. We consider the massive sulphide potential the highest, mainly because of the size and regularity of both geochemical and geophysical anomailes and the amount of massive sulphide float."

Bob Lane, Geologist, BC Geological Survey (1996):
"I am especially interested in the Ace property, which I view as a significant new discovery-one that has ramifications for exploration in Downey succession rocks.

Bob Lane, Regional Geologist, Ministry of Energy and Mines
R. Lane and K. McDonald, Geologist, BC Geological Survey (2000):
"The Ace's host rock lithology and metal content suggest an affinity to well explored gold quartz veins of the Yanks Peak and Cow Mountain areas to the north. The geological setting, style of mineralization and geochemistry suggest an analogy to the "Plutonic-associated" or "Pogo-Type" Gold vein model."

British Columbia Mineral Exploration
Review 2000
Information Circular 2001-1
Ministry of Energy and Mines (Tom Schroeter)
"In the Likely area, 75 kilometres northeast of Williams Lake, Barker Minerals Ltd. continued geological mapping and geophysical surveys over its Frank Creek and Ace volcanogenic massive sulphide showings. Although the felsic rocks at Frank Creek are different from those at the Ace showing to the northeast, the proximity of the two zones in the Barkerville Terrane enhances the potential of discovery of more VMS deposits within the belt".

R. MacDonald, Teck Exploration Ltd. (1999):
"High grade assays from the Frank Creek property visits and Barker Minerals' exploration results from the Ace and other prospects underscore the significant untapped exploration potential of the Likely area. Furthermore, we are encouraged by the variety and widespread distribution of base and precious metal mineralization over the entire property package."

Robert C. Bell, INCO Technical Services (2000)
"We appreciate the excellent cooperation that we have received from you and your team in aiding us in our review, and commend you for the highly professional manner in which you have carried out your exploration and compiled the resulting data."

4.1 Ace VMS and Vein-gold Project

The Ace property is underlain mainly by a section of metamorphosed sedimentary and volcanic rocks. The most important sedimentary types are quartz-muscovite schist, quartz-chlorite-muscovite schist, quartzite, quartz-rich schist and quartz-muscovite-biotite schist. In the eastern half of the property, garnet is common, especially in muscovite-rich and biotite-rich schist. Minor argillite, calcareous argillite and limestone are most common directly above the massive sulphide lenses. Deeper in the section on Barker Mountain and along Ishkloo Creek are minor amphibolite and biotite amphibolite.

A volcanogenic massive-sulphide environment is associated with metamorphosed felsic volcanic rocks along the trend of Doyle's boulder field, where many boulders containing gold-bearing quartz veins and gold-bearing massive sulphides have been found. Drill holes intersected a zone of metamorphosed felsic volcanic rocks up to several km long and 80 m thick. These felsic rocks are dominated by plagioclase with minor to moderately abundant quartz, muscovite and biotite.

According to very recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), the host rocks of the Ace mineralization are relatively metamorphosed and structurally deformed. They include a variety of quartz-muscovite-chloritebiotitegarnet schists, a pale plagioclase-bearing rock up to several tens of metres thick that has been previously referred to as "felsite", and minor argillite and marble. Lithogeochemical data indicate that the "felsite" in fact is compositionally similar to calc-alkaline andesite. However, due to metamorphism and recrystallization, it is not certain if it represents a true volcanic rock or a clastic sediment such as greywacke. Amphibolites which probably represent mafic volcanic rocks are also present within 1-2 km of the Ace sulphide occurrence.

At the top of the main felsic section in DDH 98-03 is a zone 0.75 m thick, containing two intersections of massive sulphide mineralization 0.20 m thick and 0.25 m thick, separated by an intersection of calcite 0.30 m thick. The massive sulphide mineralization carries anomalous concentrations of Au, Ag, Cu, Zn, Bi, Se and Mn, as determined by non-statistical consideration of the analytical data.

A few massive sulphide samples collected from trenches in this zone contain up to 2% Zn. A similar zone of semi-massive sulphides 0.35 m thick occurs in the centre of the section of felsic volcanic rocks in DDH 98-07. Massive sulphide boulders containing large concentrations of Pb and Zn occur in the Colleen Road area, but their bedrock source has not been found.

In some strongly altered footwall rocks, the host rock was replaced completely by massive, fine-grained pyrite and dark green chlorite. Others contain 20-30% disseminated, very fine-grained pyrrhotite. Many altered felsic rocks contain anomalous concentrations of base metals and precious metals, in part associated with recrystallized and/or replaced patches dominated by one or more of plagioclase, quartz, ankerite and sulphides. Anomalous concentrations increase in footwall rocks near the stratigraphic top of the main felsic section. These patterns show characteristics of footwall rocks beneath a typical VMS deposit.

The drill cores contain a few important faults up to a few metres wide. Because of the wide spacing of the drill holes, these could not be correlated between holes. The moderate to high graphite content of many of these makes them potentially significant conductors.

Most of the geophysical anomalies obtained in earlier studies have yet to be tested or explained. The main geophysical and geochemical anomaly at the western end of the main trench area is open to the west in an area that is interpreted to be underlain by felsic volcanic rocks. This extends west of the massive sulphide occurrence in DDH 98-03. Vectors in the thickness of massive sulphides, intensity of alteration and geochemical anomalies and thickness of the felsic volcanic section indicate that the area west of DDH 98-03 may contain an important exploration target. Another exploration target is indicated by a broad geophysical resistivity anomaly situated northeast of the area of drilling (Figure 4a Perry 2002). On surface, this area contains rubble of felsic volcanic rocks and abundant boulders of quartz veins containing anomalous concentrations of base and precious metals. A third important target is the elongate HLEM anomaly in the southeastern part of the project area (Conductor A, Figure 4a Perry 2002).

Delineation of the "felsite" unit led to the discovery of massive to semi-massive mineralization in the 5N Zone, an area that was extensively trenched prior to 1998. Three holes drilled into the zone in 2002 have defined the felsite package as a unit some 60-80 metres thick, possibly thickened by a series of tight to isoclinal folds. Immediately to the west in ACE-02-05 and the east in DDH-98-04, the unit breaks up into a series felsite layers, separated by weakly chloritic quartz-muscovite schists. In the 16S Area, ACE-02-01 was collared to test coincident max-min conductor and subtle gravity anomaly on line 16S. Wild indicated that both the conductor and gravity were caused by a 6.5-metre thickness of felsite with up to 10% pyrite and pyrrhotite, however follow-up geophysical surveys suggest that this hole may not have tested the intended geophysical anomalies, as the zone may be faulted off. The follow-up geophysics defined potentially significant targets situated to the west and also possible faulted off extensions of the favorable horizon to the east (Walcott, 2002). Drill hole ACE-02-02, was collared immediately above the felsite and tested the footwall near a large ground magnetic high to the south. That anomaly was not adequately explained in the core. (Wild, 2002)

The felsite unit is reportedly made up of up to 80% plagioclase (Payne, 1998), although at least one sample was composed of mainly albite (Walus, 1997). The fine-grained nature of the rock, its intimate relationship to a series of thin marble bands, and almost regional extent suggests that it may be an exhalative horizon and an excellent target horizon for VMS deposits. Sulphide mineralization in 98-3, ACE-02-03, and ACE-02-04 is located near the structural top of the unit.

Gold-bearing quartz veins in this area occur in two main modes. The first are as early veins that were deformed strongly with the enclosing host rock. The second are as late crosscutting veins, many of which trend northeasterly. Many of the quartz veins occur in the same general area as the felsic rocks and boulders of massive sulphide.

The Ace project area has significant similarities to the Intrusion Related Gold System model setting as advanced by Lang, Baker, Hart and Mortenson (Society of Economic Geologists Newsletter #40, January, 2000), as it contains geology and mineralogy similar to that of the Yukon and Alaskan settings that host a variety of intrusion related gold deposits (Lane, 2000) such as Fort Knox (158 Mt@ 0.83 g/t Au; Bakke, 1995), Brewery Creek (13.3 Mt @1.44 g/t Au; Diment and Craig, 1998), Dublin Gulch (50.3 Mt @ 0,93 g/t Au; Northern Miner, 1997), True North (16.8 Mt @ 2.5 g/t Au; Harris and Gorton). The newest and highest-grade discovery in the belt is the Pogo deposit (Smith et al., 1999; 9.98 Mt @ 0.52 oz/t Au). According to independent geologists of Teck Exploration Ltd., who expressed that they wish to continue close monitoring of the Barker projects, it is clear from the data that multiple mineralizing episodes are present throughout the property (pers. comm., R. Macdonald, Teck Exploration Ltd., December 1999).

4.2 Frank Creek VMS Project

The Frank Creek area contains an important massive sulphide occurrence situated near the stratigraphic top of a fragmental, felsic volcanic rocks or feldspathic arkose. This overlies, in order, a section of black argillite and siltstone and an intermediate to mafic volcanic sequence of flows and fine fragmental rocks. Associated with the massive sulphide zone is a Cu-rich zone of stringer and replacement mineralization. The stratigraphic section is on the overturned limb of one of a set of major, southwesterly verging, F2 folds. All the rocks are metamorphosed, which has inhibited interpretation. The discovery of pillow structures in mafic volcanic rocks in Frank Creek indicates a sea-floor subaqueous environment, thereby enhancing the potential for further discoveries of massive sulphide deposits in this belt of rocks. The association of the Frank Creek massive sulphide and the Ace massive and stringer sulphide mineralization to volcanic rocks enhances the potential for discovery of additional VMS mineralization not only the Goose Range, but throughout the entire Barkerville terrane.

According to recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), original stratigraphic relations and rock textures are much better preserved here than at the Ace VMS environment. Host rocks include quartz-eye-bearing sandstones, siltstones, shales, and minor mafic sills. Work is in progress to determine if the quartz-eye-bearing sandstones represent volcanic crystal tuffs or clastic sediments. Pillow basalts are present 1-2 km south of the Frank Creek sulphide occurrence and basaltic volcaniclastic beds also outcrop in the area, although their stratigraphic position relative to the mineralization is currently uncertain. Nonetheless, the presence of extrusive rocks raises the possibility that mafic volcanism was occurring near the time of Frank Creek mineralization.

The massive sulphide outcrop, spatially related boulders and the Cu-rich stringer and replacement zone contain anomalous to potentially economic concentrations of Cu, Pb, Zn and Ag. Many samples are anomalous in Au and some contain anomalous concentrations of As, Bi, Cd, Hg, Sb, Tl and Mo. Concentrations of As, Bi, Hg, Mo and Sb also are anomalous in stream-sediment samples near other airborne geophysical anomalies. The HLEM survey suggests the presence of another massive sulphide target stratigraphically a few tens of metres above the massive sulphide outcrop. During 2001-2002, trenching and drilling at the Frank Creek VMS project were largely successful in exposing the mineralized zone over a strike length of approximately 425 metres. Based on the results to date, further trenching and drilling are recommended. (Perry, 2002)

According to recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), the Ace and Frank Creek sulphides occur within the late Precambrian to lower Paleozoic Snowshoe Group of the Barkerville terrane. Based on regional geology, the paleotectonic setting of the Snowshoe Group is generally considered to have been a marine continental shelf which accumulated sandy to muddy clastic sediments and locally mafic to andesitic volcanic rocks. The Cu-Zn-Pb-bearing massive sulphides at Ace and Frank Creek are at least partly hosted by sedimentary rocks. Relative to typical VMS deposits, the massive sulphides are locally enriched in trace metals such as Sb, As, Bi, Hg, In and Sn, which is consistent with derivation of at least some of the metals from hydrothermal leaching of an underlying sedimentary sequence. Possible models for the setting of the massive sulphides, at this early stage of exploration, range from a sediment-dominated continental margin with local, mainly mafic volcanism along extensional faults, to a deeper-water Besshi-type setting, to a sediment-covered spreading axis. The proposed stratigraphic and lithogeochemical studies should serve to define the paleotectonic setting of the mineralization, the most favourable host lithologies, and the distribution of hydrothermal alteration, which in combination will provide a more focused exploration model for the area.

According to Wild (2002), former Chief Geologist of Goldstream Mine, near Revelstoke, B. C., the geological setting, mineralization and host rocks are all remarkably similar to the Goldstream Mine mineral deposit, which produced more than 2 million tonnes of ore at a grade in excess of 4.0% copper and 2.2% zinc.

The Big Gulp showing (F-4 target area) contains anomalous concentrations of Zn in strongly altered intermediate to mafic volcaniclastic rocks associated with a series of east-west-trending HLEM anomalies (F-4 in Figures 8a and 9 Perry 2002). This may be indicative of a stringer zone associated with a massive sulphide deposit.

4.3 SCR Project

A significant, new VMS prospect has been identified with the discovery of semi-massive sulphide and stringer sulphide mineralization in float and bedrock of altered intermediate to mafic volcanic rocks of the Sellers Creek Road area. Sulphide minerals include pyrite, pyrrhotite, chalcopyrite, sphalerite and galena. The geophysical surveys successfully outlined significant, coincident magnetic and conductive anomalies (HLEM) near the discovery area and near the area of Cu, Pb and Zn soil anomalies. The presence of volcanogenic massive sulphides at the Ace, Frank Creek and SCR properties shows that potential exists for VMS deposits across the entire width of the Barkerville terrane (Payne, 2001; BC Assessment Report).

4.4 Quesnel Platinum Project

Reconnaissance geology and thin section petrology confirmed the occurrences and character of mafic and ultramafic rocks. These are probable sources of some of the PGMs in the placers of the Quesnel River and its tributaries. Several samples from hematite-altered amygdaloidal mafic flow rocks are anomalous in Cu, Ag and Hg. A recent BC Geological Survey study confirmed the presence of PGM's in the Quesnel River and some of its tributaries draining Barker's Quesnel Platinum Project areas, especially Black Bear Creek (Levson *et al*, 2002)

5.0 Recommendations

Various geological and mining professionals have been involved in the Barker Minerals Projects or have reviewed the Company's exploration work and made field examinations of the active projects. Dr. Bruce Perry has consolidated the recommendations with which he is in agreement, has added his own recommendations and has added emphasis, in some cases, to the recommendations of others.

5.1 Ace VMS and Vein-gold Project

Geological mapping should continue in order to improve understanding of the regional structure and the local geology of areas of felsic volcanic rocks that have not yet been examined. This additional mapping should be integrated with that being done between the Ace and Frank Creek areas by Ferri and others of the B.C. Geological Survey. Wild (2002) recommended extensive whole rock analysis to better characterize the felsite unit. Independent geological consultants of Strathcona Mineral Services (Toronto, Canada) toured the Ace Project and after inspecting core from the drill programs of 1998 and 2002 recommended further work including delineation of the felsite unit through mapping, soil geochemistry and geophysical surveys, followed by trenching where possible and drilling of targets which are selected by the combination of magnetic, MaxMin and gravity geophysical surveys. They are in agreement with Wild (2002) and Perry (2002) that an effort should be made to determine the origin of the felsite, as this has a bearing on the style of massive sulphide deposits that may exist in relation to this unit on the Company's property.

Barker Minerals has very recently enlisted the services of Dr. Tim Barrett and Dr. Wallace MacLean of Ore Systems Consulting (Toronto, Canada) to determine the stratigraphic and alteration relations of the rock sequences hosting the massive sulphide mineralization on the Ace and Frank Creek properties using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography. Ore Systems Consulting has previously worked on numerous volcanic-associated massive sulphide deposits, including those in the Noranda and Matagami camps of Quebec, the Timmins area of Ontario, the Cordillera of British Columbia, Sweden, Wales, Portugal and the Philippines, and have published numerous research papers on these areas, Perry (2002) recommends that they determine the paleotectonic and paleoseafloor setting of the mineralization in each area, as this will have a direct bearing on the VMS model used during exploration. Perry (2002) also recommends work to determine the chemical identification of the main mineral host rock units at the Ace and Frank Creek Projects in order to trace these favorable units laterally during exploration. The quantitative determination of the degree of alteration is also recommended in order to determine the most intense portions of the hydrothermal systems on each property, as those areas containing a combination of favorable stratigraphy and strong hydrothermal alteration have the greatest potential for economic VMS mineralization. The results should be combined with known geophysical and soil-geochemistry anomalies established both by earlier surveys and potential 2003 surveys in order to help select high-priority drilling targets.

A study should be undertaken to better understand the genesis of the highgrade gold/bismuth/tungsten/telluride mineralization present in the very numerous float boulders found on the property. This will assist in developing specific strategies for exploration for the bedrock source(s) of these boulders and similar occurrences.

The HLEM and geochemical surveys should be extended to the west and to the east of the present surveys along the trend of the felsic volcanic rocks. Similar surveys should be conducted over the belt of felsic volcanic rocks north of Little River. The main HLEM electromagnetic anomaly discovered during 2000 has been tested by two gravity profiles which identified excess mass on both test lines that was co-incident with previously defined HLEM (Maxmin) anomalies. Similar gravity profiles should be run across the core of the altered felsic volcanic rocks defined in the 1998 drill program.

An IP survey should be conducted in order to verify the low resistivity e-scan-indicated target.

Additional detailed ground magnetic surveys are recommended in order to continue to identify drill targets which could be possible bedrock sources of the numerous clusters of magnetic pyrrhotite, gold-rich quartz sulphide boulders occurring on the Ace property.

Trenching should be conducted on anomalous zones where the thickness of the glacial overburden allows.

Diamond drilling should test targets identified by the existing geological, geophysical, geochemical and trenching results and by additional work that will be conducted in the first part of the 2003 field season. Further compilation work is required for the 5N Zone to determine if further drilling is required there next, or whether further surveys are required in preparation for potential drilling. According to Wild (2002), to the east, soil sampling, is warranted on some of the recently established grid on the east half of the project area. Also to the east, a coincident max-min conductor and subtle gravity anomaly on lines 3800S to 4700S should be followed up by extending the grid and extending the gravity coverage to adjacent lines. Drilling should be planned on the most promising part of the anomaly (Wild, 2002). Similarly, Walcott concluded that the residual gravity anomalies obtained in relation to EM conductors here be investigated by drilling (Walcott, 2002).

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

5.2 Frank Creek VMS Project

Geological mapping should continue in order to help determine the stratigraphic and structural setting of the mineralization. This should be integrated with that being done by Ferri of the B.C. Geological Survey between the Ace and Frank Creek areas. Independent geological consultants of Strathcona Mineral Services (Toronto, Canada) toured the Frank Creek Project and after inspecting core from the drill program of 2002 recommended further work including establishing survey grids, mapping and soil sampling and geophysical surveys similar to those recommended for the Ace Project, followed by trenching and drilling. They concluded that there is still the possibility that massive sulphide mineralization exists on the property and that more work is warranted on the remaining targets.

Barker Minerals has enlisted the services of Dr. Tim Barrett and Dr. Wallace MacLean of Ore Systems Consulting (Toronto, Canada) to determine the stratigraphic and alteration relations of the rock sequences hosting the massive sulphide mineralization on the Ace and Frank Creek properties using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography. Ore Systems Consulting has previously worked on numerous volcanic-associated massive sulphide deposits, including those in the Noranda and Matagami camps of Quebec, the Timmins area of Ontario, the Cordillera of British Columbia, Sweden, Wales, Portugal and the Philippines, and have published numerous research papers on these areas, Perry (2002) recommends that they, based on their worldwide experience and advance level of geological expertise in the field of volcanogenic massive sulphide deposits and models, determine the paleotectonic and paleoseafloor setting of the mineralization in each area, as this will have a direct bearing on the VMS model used during exploration. Perry (2002) also recommends work to determine the chemical identification of the main mineral host rock units at the Ace and Frank Creek Projects in order to trace these favorable units laterally during exploration. The quantitative determination of the degree of alteration is also recommended in order to determine the most intense portions of the hydrothermal systems on each property, as those areas containing a combination of favorable stratigraphy and strong hydrothermal alteration have the greatest potential for economic VMS mineralization. The results should be combined with known geophysical and soil-geochemistry anomalies established both by earlier surveys and potential 2003 surveys in order to select high-priority drilling targets.

The HLEM survey should be continued in areas of the grids not yet covered, especially to trace continuations of known anomalies that are suggestive of the presence of massive sulphide deposits. Additional preliminary gravity survey lines should be run across the main anomalies to more accurately define massive sulphide targets. Where topography and glacial cover are permissive, trenches should be excavated into prospective targets resultant from these studies. VLF geophysical surveys should be used to trace mineralization once it has been located in bedrock.

Geochemical soil surveys should be conducted over the strike extensions of known mineralized trends and also over specific target areas in order to identify base and/or precious metal soil anomalies, which may be indicative of economic massive sulphide mineralization targets in the local bedrock.

The most prospective anomalies as defined by geology, geophysics, geochemistry and trenching should be tested by diamond drilling.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

5.3 SCR Project

Geological mapping should continue in order to help determine the stratigraphic and structural setting of the mineralization. This should be integrated with that being done by Ferri of the B.C. Geological Survey between the Ace and Frank Creek areas.

Lithogeochemistry studies should be continued in order to determine the stratigraphic and alteration relations of the rock sequences hosting the known mineralization, using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography.

The HLEM survey should be continued to the east and southeast of the present grid to determine the extent of the anomaly in those directions. Where permissive, trenching should be conducted on defined target areas.

Soil geochemistry should be conducted over the grid and should be extended to the east to cover the possible extensions of the anomalous zone as defined by geology and coincident magnetic and HLEM anomalies.

Diamond drilling should test the most prospective anomalies as defined by the geology, geophysics, geochemistry and trenching.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

5.4 Peacock Showing

A mineral occurrence was found by a BC Geological Survey geologist during his local geological mapping work may be the previously known Peacock mineral showing. The Company should further expose and sample this mineralization. This occurrence has been interpreted by government geologists (Ferri, 2001) to be Besshi-type VMS mineralization. All available previous exploration results should be compiled, interpreted and, if then warranted, be followed up with an initial program designed to identify and develop drill targets having economic VMS and/or gold/silver potential.

5.5 Blackbear Project Area

Since it appears that interest in gold is increasing as its price has risen substantially during the previous year, the Company should begin to investigate this project area, which hosts the former Providence Mine, from which a previous operator's shipment of selected ore from the No. 2 zone assayed 3343 grams of silver per tonne, 45.7% lead, 0.11% zinc and 4.9 grams of gold per tonne (all the above from BC MinFile 093A 003). One grab sample (# 11-07-98-59; Barker Minerals) of outcropping mineralization contained 52% Pb, 0.03% Zn, 142 oz Ag / ton and 0.081oz Au/ton. (Payne, 1999; BC Assessment Report). All previous exploration results available should be compiled, interpreted and, if then warranted, be followed up with an initial program designed to identify and develop drill targets having economic VMS and/or gold/silver potential.

5.6 Cariboo Prospect and Other Areas

Detailed geological mapping should be continued in the Cariboo prospect area in order to help determine the extent of deformation and in order to explore for targets of Zn+Pb replacement deposits along strike of the known zones. Compilation of all relevant data and limited diamond drilling is recommended in order to confirm the previous operator's drilling and in order to further define and investigate the size and economic potential of this deposit, which is open in both directions along strike and at depth. Despite low gold concentrations found there to date, further prospecting in the near vicinity and exploratory investigation of the Foster zone for its gold potential is recommended in light of its position within the Pleasant Valley Thrust and the intensity of sulphides in the zone and their localization at a junction of two large scale structures.

Elsewhere in the eastern half of the property, detailed mapping and follow-up geochemical sampling and geophysical surveys should be performed in areas of significant, multi-element stream-sediment anomalies. Areas of particular interest are the upper reaches of the Sellers Creek and Grain Creek drainage basins.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

5.7 Quesnel Platinum Project

Stream and soil sampling programs should be conducted in drainages associated with known placer occurrences of PGE minerals, especially in areas of mafic and ultramafic rocks that may be sources of such minerals. Studies should be done in order to characterize the assemblages of PGEs, which could help determine the environment of their origin. Geological mapping should be conducted in areas of mafic and ultramafic rocks in order to better understand the nature of these rocks and their possible association with PGE minerals. Prospecting should continue in drainages that contain known placer Pt deposits or anomalous concentrations of PGEs.

The zone of large copper concentrations in the Mag claims should be studied in more detail geologically, geochemically and geophysically. Other zones of coincident geochemical and geophysical anomalies should be examined once data from previous reports are compiled and interpreted. Some of these areas will require new grids for geophysical and geochemical surveys.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

6.0 Budgets

The estimated costs of the Phase 1 and Phase 2 programs are $1,780,000 and $4,000,000 respectively (Perry 2002).

7.0 Certificate or Qualifications

Report was prepared by Louis E. Doyle, Prospector and President of Barker Minerals Ltd. All data compiled from company technical reports, which are attached as appendices.

Barker Minerals Ltd.
2002 Sample Descriptions

Sample #	Target	Description	Location
02-111	SCR	Large angular float (6ft deep) stringer type, siliceous, PY, SPH?, CPY? bands, greenish color	B/L0+00-15ms
02-111A	SCR	Large angular float (6ft deep) stringer type, siliceous, PY, SPH?, CPY? bands, greenish color	B/L0+00-15ms
02-112	SCR	(8 feet deep) Float, siliceous, felsic? DIS PY, PO, minor CPY, (light bluish)	B/L0+00-0T00N
02-112A	SCR	(8 feet deep) Float, siliceous, felsic? DIS PY, PO, minor CPY, (light bluish)	B/L0+00-0+00N
02-113	SCR	Local float, dark green, PY, stringer, siliceous	0+50N-0+50E
02-113A	SCR	Local float, dark green, PY, stringer, siliceous	0+50N-0+50E
02-114	SCR	Local float, bluish GN/SPH/CPY DIS and wispy, white mica, PY, slightly magnetic	L6s-4+00E
02-114A	SCR	Local float, bluish GN/SPH/CPY DIS and wispy, white mica, PY, slightly magnetic	L6s-4+00E
02-115	SCR	Local float rusty boulder with some visible GN	L6s-4+00E
02-115A	SCR	Local float rusty boulder with some visible GN	L6s-4+00E
02-116	SCR	Local float, blue siliceous rock, Felsic? DIS, PY,PO,CPY, SPH? Sericite, GN	L6s-3+75E
02-116A	SCR	Local float, blue siliceous rock, Felsic? DIS, PY,PO,CPY, SPH? Sericite, GN	L6s-3+75E
02-117	SCR	Bedrock - carbonaceous, bluish, siltstone, some folication, fine grained	L4+90s-4+55E
02-117A	SCR	Bedrock - carbonaceous, bluish, siltstone, some folication, fine grained	L4+90s-4+55E
02-118	SCR	Bedrock SPH, GN, bluish, carbonate, alteration, minor carbonaceous material on foliation planes	L4+90s-4+55E
02-119	SCR	Bluish siltstone, some sulphides, weathering	L4+90s-4+55E
02-119A	SCR	Bluish siltstone, some sulphides, weathering	L4+90S-4+55E
02-120	SCR	Local float semi massive sulphide, football size, 6ft down, PY, PO?, Magnetic? CPY, SPH, GN? starting to oxidize	L5S-4+65E
02-120A	SCR	Local float semi massive sulphide, football size, 6ft down, PY, PO?, Magnetic? CPY, SPH, GN? starting to oxidize	L5S-4+65E
02-121	SCR	Bedrock repeat sample: Dark bluish/black siltstone, carbonaceous, some GN,SPH wisps?	4+75S4+75E
02-121A	SCR	Bedrock repeat sample: Dark bluish/black siltstone, carbonaceous, some GN,SPH wisps?	4+75S4+75E
02-122	SCR	Float 11ft down in till, felsic? Stringer type with some fresh sulphides	4+75S4+75E
02-122A	SCR	Float 11ft down in till, felsic? Stringer type with some fresh sulphides	4+75S4+75E
02-123	SCR	Float 8ft down in till, rusty oxidized Q-eyes/felsic clasts?	L5+15E-4+50S

Barker Minerals Ltd.
2002 Sample Descriptions

Sample #	Target	Description	Location
02-123A	SCR	Float 8ft down in till, rusty oxidized Q-eyes/felsic clasts?	L5+15E-4+50S
02-124	SCR	Float 9ft down in till, felsic volcanic?Q-eyes, feldspars	L5+15E-4+50S
02-124A	SCR	Float 9ft down in till, felsic volcanic?Q-eyes, feldspars	L5+15E-4+50S
02-125	SCR	Sub outcrop, cherty siltstone/mudstone SPH?	L5s-4+65E
02-125A	SCR	Sub outcrop, cherty siltstone/mudstone SPH?	L5s-4+65E
02-026	SCR	(Oxidizing angular 6ft down float)is eliminated to semi-massive PY,PO?,CPY,SPH? Chlorite,sericitic(slightly magnetic)	TH-L5+50E4+75S
02-027	SCR	15ft down float, angular local float, felsic, Q-eyes, oxidized sulphides, most of till is the rock type (non magnetic). Panned the till, concentrate is high PY & SPH.	TH L+50E 4+75S
02-028	SCR	Local float? slightly magnetic 8ft down in till, serictic, oxidizing, PY, CPY, SPH? angular, schist, panned the till conentrate, high in PY & SPH.	TH-L5+50E4+50S
02-029	SCR	Local float 6ft down high SPH/GN in schistose mostly oxidized,angular	TH-L6+00S4+00E
02-031	F-1	Massive Graphite?blocky shiny particles?metalic?gold?	Frank 02TR on L 55
02-035	F-7	Float from 30ft down near bedrock, semi-massive to massive bands(SPH?)PY,within silicieous host rock	F-7 L59+75N24+25W
02-036	F-1	Q-eye schist,oxidized sulphides?Near max/min conductor	F-1 TR L 55 by SWB sample #1
02-037	F-1	Large Q-eyes, sericite?schisty,rusty,feldspars?	F-1 L55 TR by SW #2 sample westend
02-038	F-1	Local float layered sulphide PY local near max min conductor	F-1 L56N 16E TR Westend float on B/R
02-039	F-1	Out crop, sericite schist dark Q-eyes, rusty?	F-1 L 56N/G ETR West end O/C

Till Samples

Sample #	Target	Description	Location
02-040	SCR	Till, 10ft depth, no bedrock	TH 02-02 10ft down
02-041	SCR	Till, 10 ft depth, no bedrock	TH 02-07 10ft down
02-042	SCR	Till, 4 ft depth, no bedrock	TH 02-07 4ft down
02-043	SCR	Till, 9ft depth	TH- 02-13 9ft down
02-044	SCR	Till, 15ft depth	TH-02-13
02-045	SCR	Till, 12ft depth	TH L 5+00E 3+25S
02-046	SCR	Till, 17ft depth, darker	TH L 5+00E 3+25S
02-047	SCR	Till, 15ft depth	L5+50E 4+75S
02-048	SCR	Till, 8ft depth	L5+50E 4+50S
02-049	SCR	Till, 11ft depth	L5+75E 4+50S
02-050	SCR	Till, 11ft depth	L6+50E 3+50S

BARKER MINERALS LTD.

SCR Project
2002 Samples

LEGEND

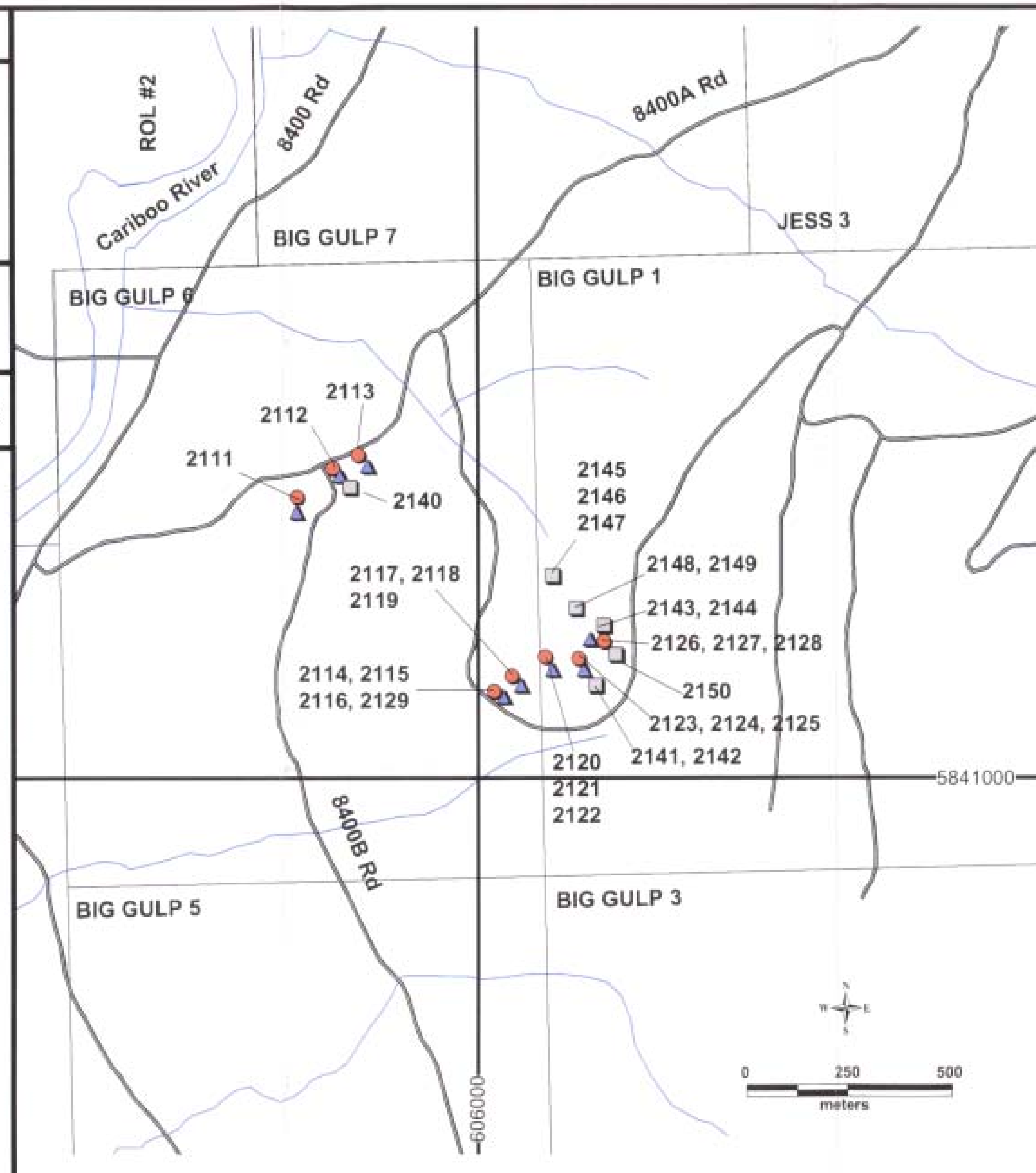
● Rock sample site ▲ Test-holes ☐ Till sample site

April 2003

Author: A. Doyle

Sample Results

Sample #	Gold-Au (ppb)	Silver-Ag (ppm)	Copper-Cu (ppm)	Lead-Pb (ppm)	Zinc-Zn (ppm)
Rock Samples					
2111	20	3	1985	30	30
2112	4	1	82	22	84
2113	2	1	85	3	36
2114	14	1	248	2130	3170
2115	168	56	1960	64000	173000
2116	6	3	819	3630	6820
2117	4	1	27	266	1080
2118	4	1	132	24	342
2119	4	1	252	23	248
2120	12	1	2150	106	40
2121	2	1	102	21	204
2122	4	1	510	12	216
2123	4	1	5	9	24
2124	2	1	6	22	40
2125	2	1	67	12	186
2126	834	1	279	69	169
2127	17	1	9	45	51
2128	2	1	460	359	97
2129	56	43	124	60700	50000
Till Samples					
2140	5	1	78	90	205
2141	14	1	65	43	166
2142	11	1	65	66	188
2143	8	1	52	47	130
2144	10	1	51	41	126
2145	18	1	54	43	141
2146	25	1	59	48	144
2147	14	1	43	39	116
2148	15	1	54	47	146
2149	17	1	55	60	161
2150	18	1	47	45	130



BARKER MINERALS LTD.

Frank Creek Project
2002 Samples

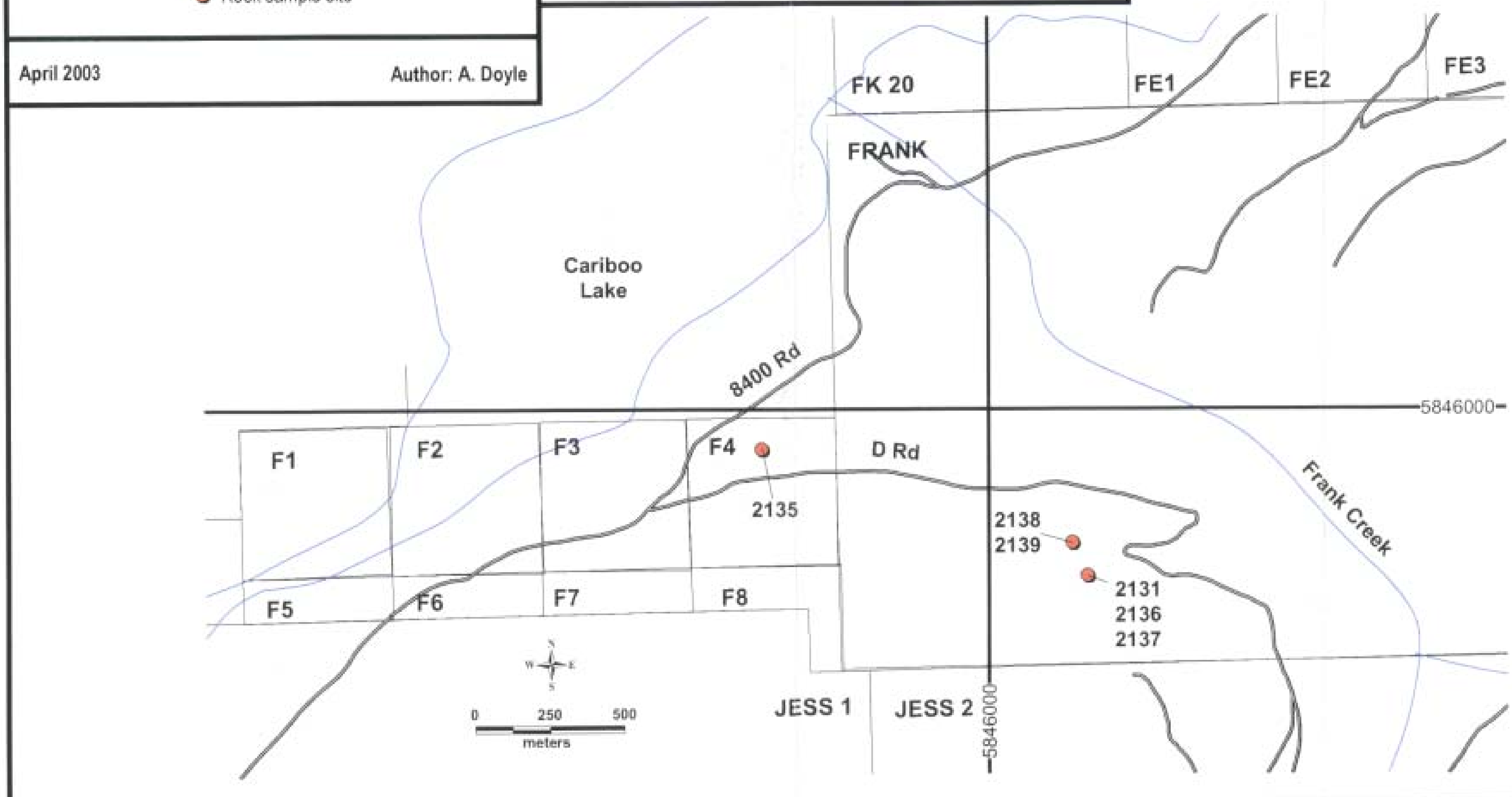
LEGEND

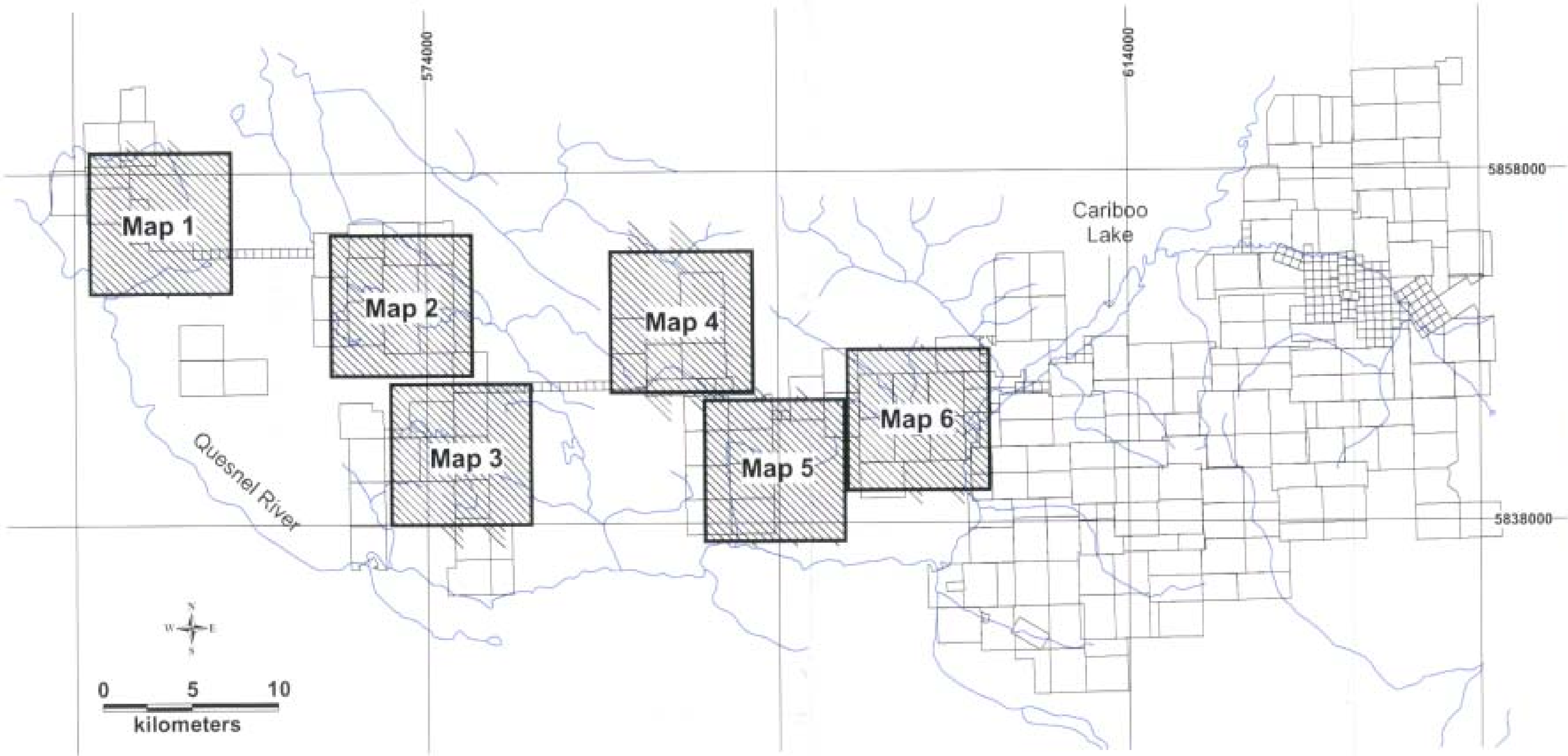
● Rock sample site

April 2003 Author: A. Doyle

Sample Results

Sample #	Gold-Au (ppb)	Silver-Ag (ppm)	Copper-Cu (ppm)	Lead-Pb (ppm)	Zinc-Zn (ppm)
Rock Samples					
2131	22	1	34	13	98
2135	69	33	3540	2320	1341
2136	7	1	96	107	276
2137	2	1	34	52	148
2138	11	11	285	151	129
2139	1	1	32	22	73



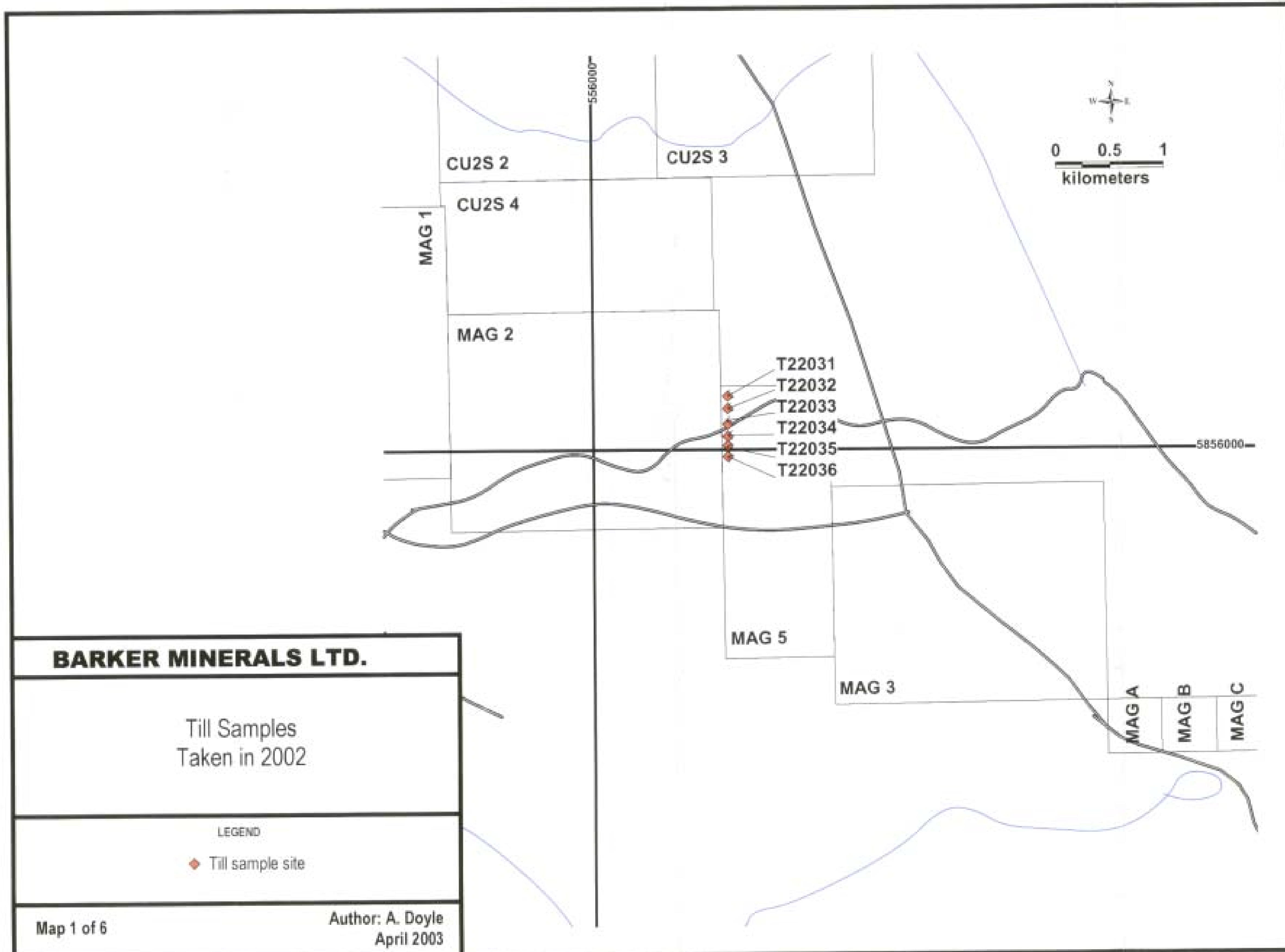


BARKER MINERALS LTD.

Index Map of
Till Samples Taken
In 2002

Index Map

Author: A. Doyle
April 2003

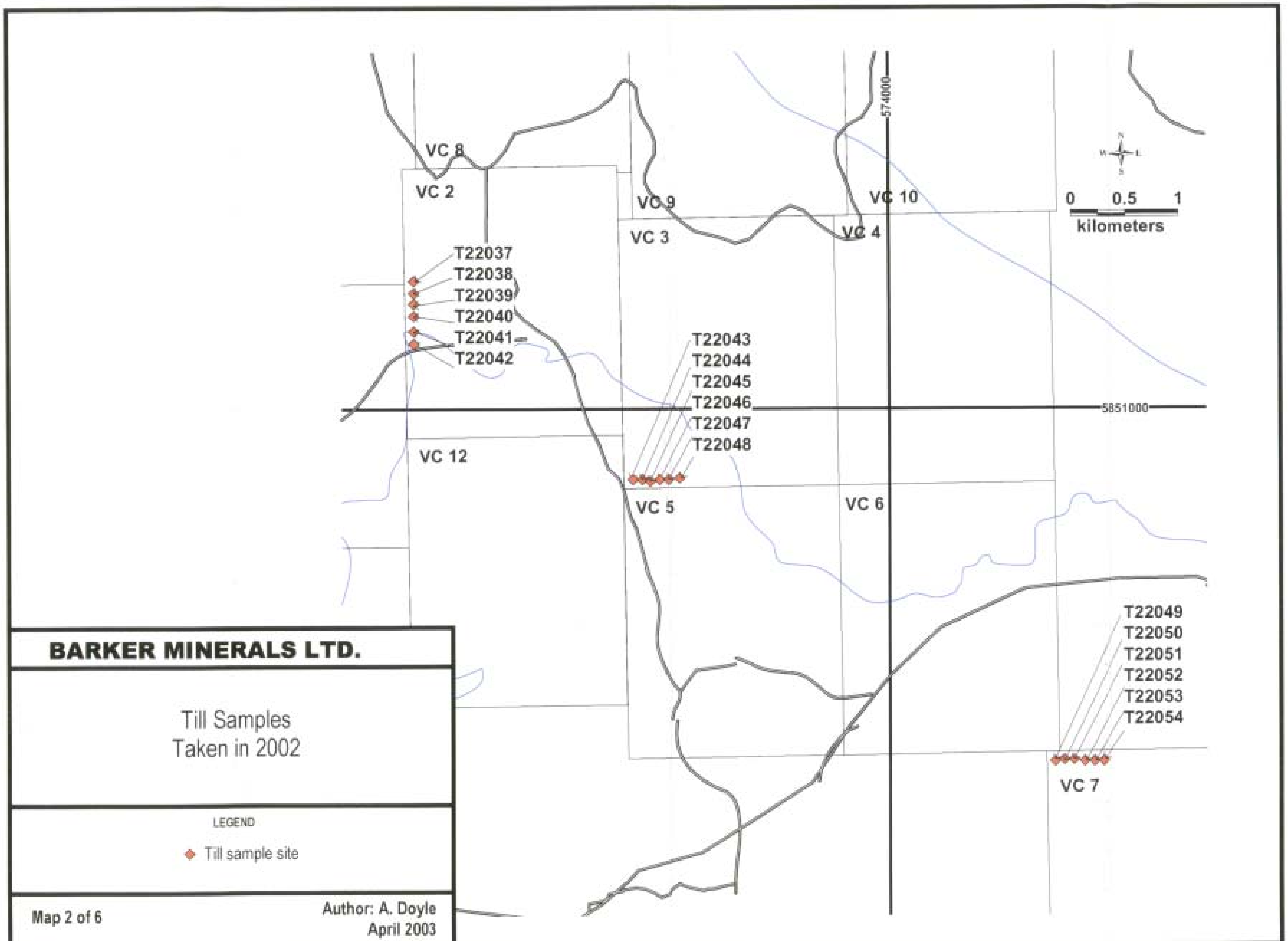


BARKER MINERALS LTD.

Till Samples
Taken in 2002

Map 1 of 6

Author: A. Doyle
April 2003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5851000
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VC 5
VC 6
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T22050
T22051
T22052
T22053
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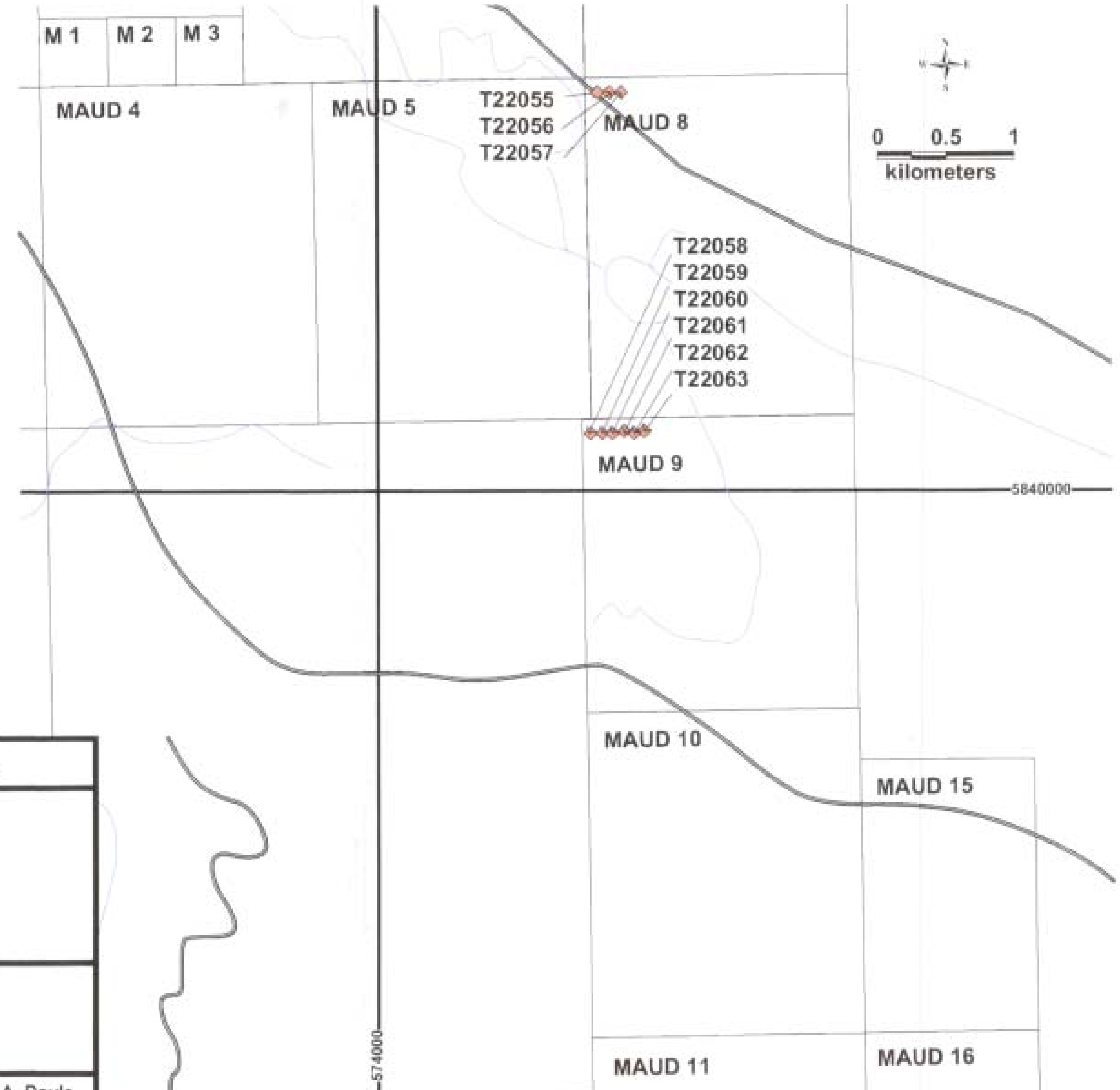
BARKER MINERALS LTD.
Till Samples
Taken in 2002
LEGEND
Till sample site
Map 2 of 6
Author: A. Doyle
April 2003



BARKER MINERALS LTD.

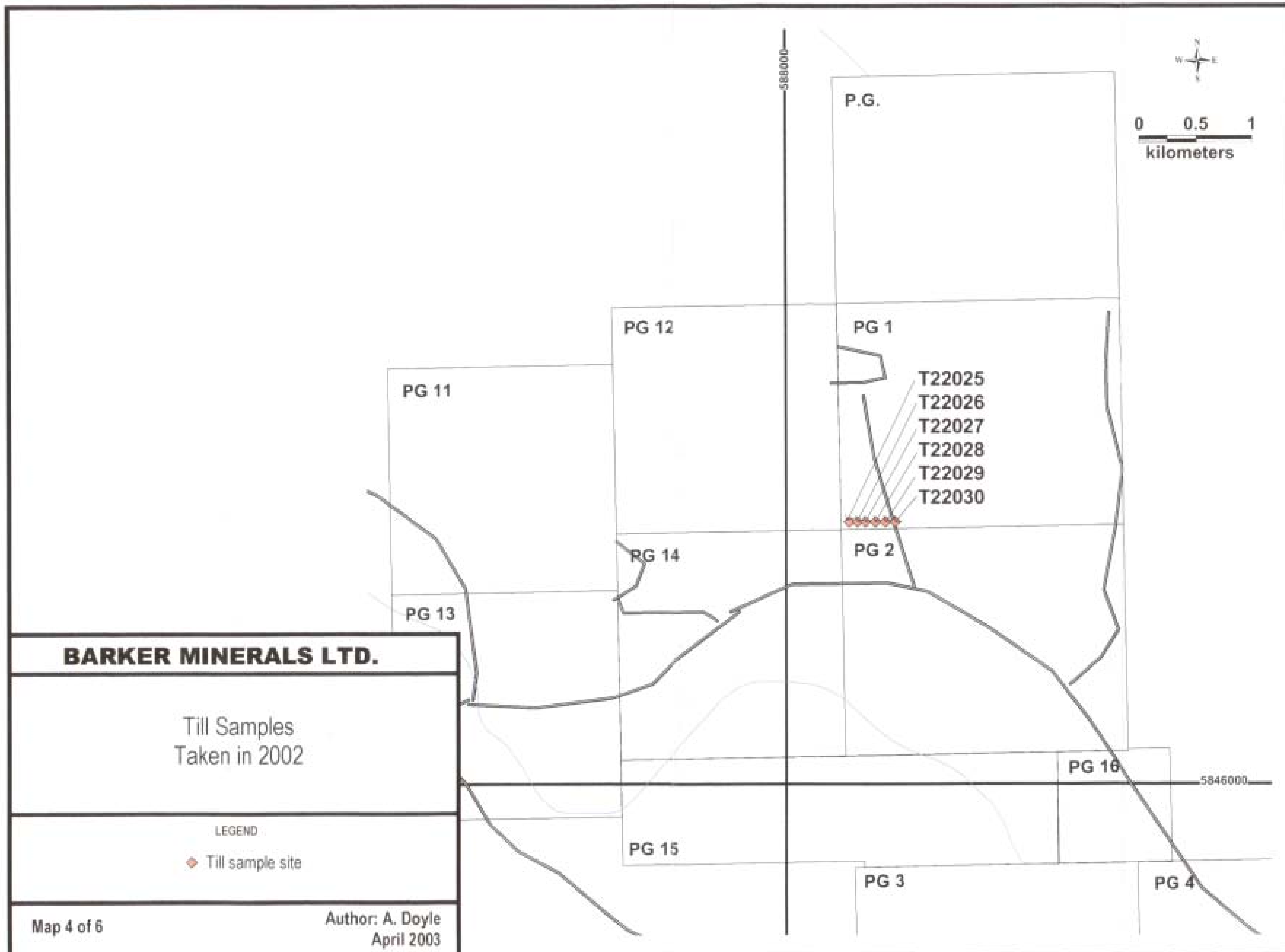
Till Samples
Taken in 2002

LEGEND

◆ Till sample site

Map 3 of 6

Author: A. Doyle
April 2003


BARKER MINERALS LTD.
Till Samples
Taken in 2002
LEGEND
Till sample site
Map 4 of 6
Author: A. Doyle
April 2003
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588000
5846000
P.G.
PG 1
PG 2
PG 3
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PG 13
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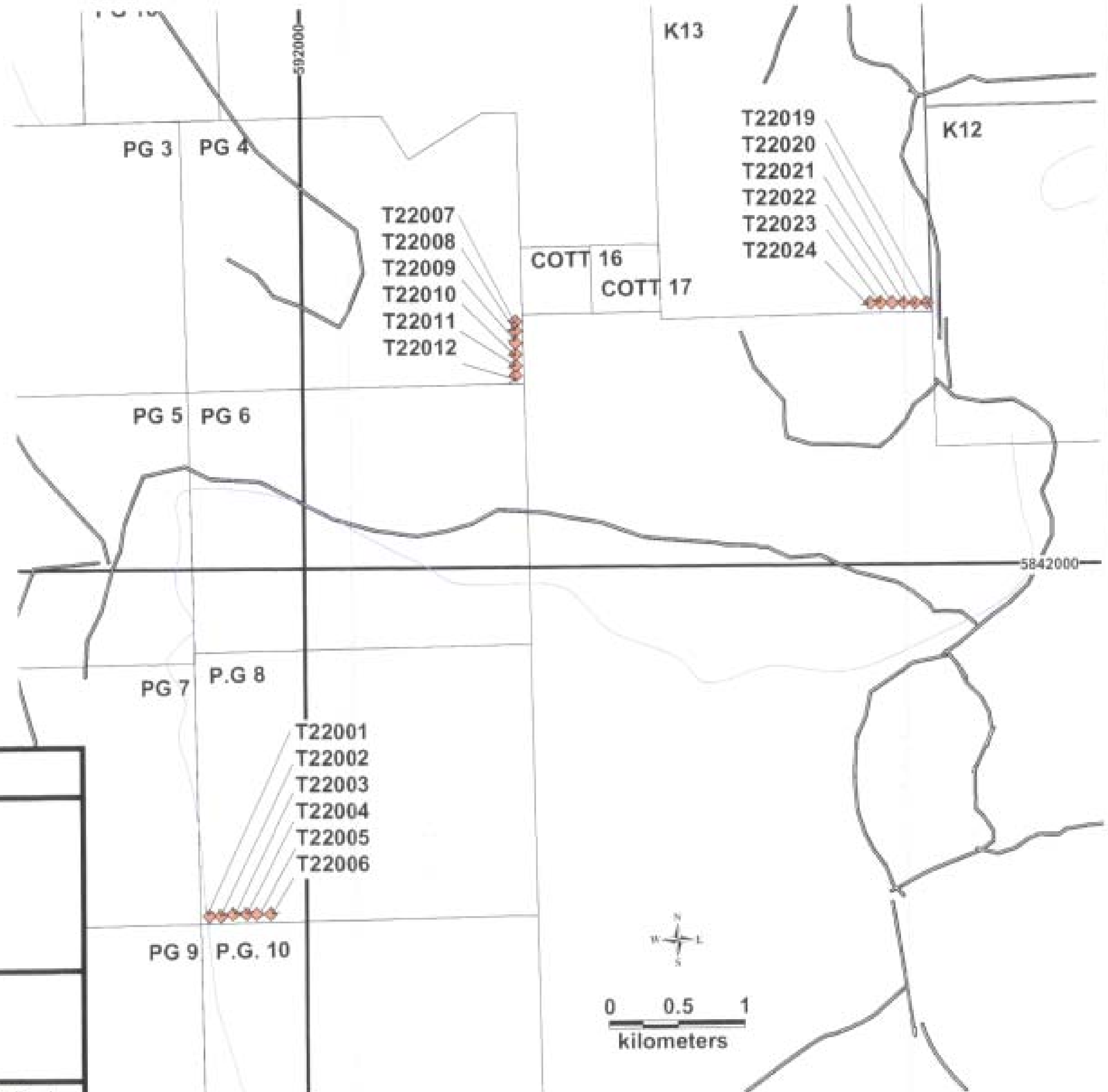


BARKER MINERALS LTD.

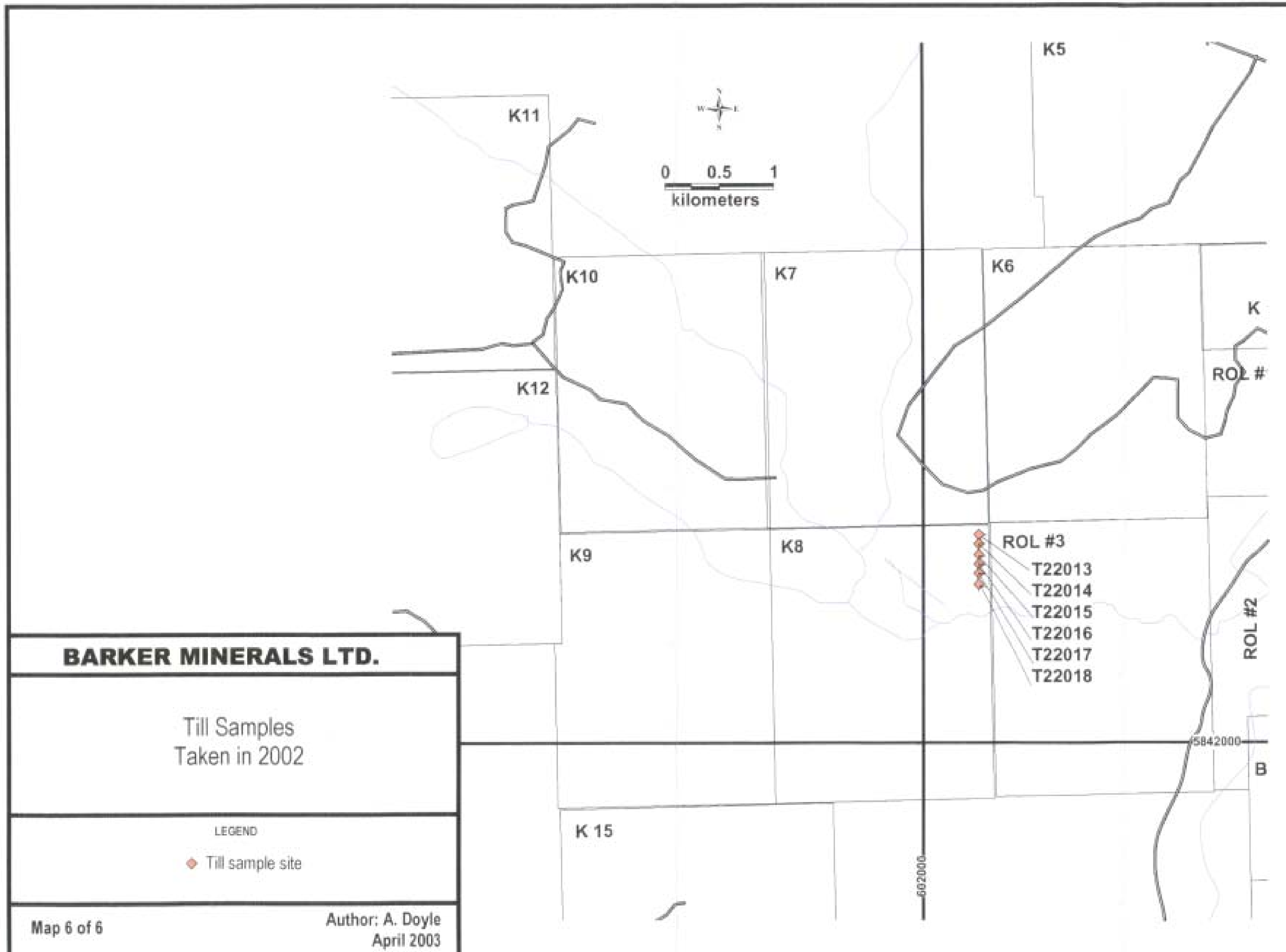
Till Samples
Taken in 2002

LEGEND

◆ Till sample site

Map 5 of 6

Author: A. Doyle
April 2003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T22017
T22018
ROL #2
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B
K 15
602000
BARKER MINERALS LTD.
Till Samples
Taken in 2002
LEGEND
Till sample site
Map 6 of 6
Author: A. Doyle
April 2003

(B) Technical

Trenching

Aaron Doyle - Operator

• 33.5 days @ $250.00/day wages	$ 8,375.00
• 33.5 days @ $90.00/day room & board	$ 3,015.00
Total	**$11,390.00**

Aaron Doyle - Swamper

• 28 days @ $200.00/day wages	$ 5,600.00
• 28 days @ $90.00/day room & board	$ 2,520.00
• 28 days @ $100.00/ day vehicle & gas	$ 2,800.00
Total	**$10,920.00**

Jim Doyle – Swamper

• 33.5 days @ $150.00/day wages	$ 5,025.00
• 33.5 days @ $90.00/day room & board	$ 3,015.00
Total	**$ 8,040.00**

Back Hoe

• 32.5 days @ $600.00/day	**$19,500.00**

Bob Westran - Rylant Construction

• 23 days @ $1,187.05/day wages	$27,302.13
• 23 days @ $90.00/day room & board	$ 2,070.00
• 23 days @ $100.00/ day vehicle & gas	$ 2,300.00
Total	**$31,672.13**

Rudy Geddert

• 8 days @ $561.75/ day wages	$ 4,494.00
• 8 days @ $100.00/ day vehicle and gas	$ 800.00
• 8 days @ $90.00/ day room and board	$ 720.00
Total	**$ 6,014.00**

KCD Contracting

• Invoiced	**$ 711.56**

Ray Savidan

• 6.5 days @ $840.26/day wages	$ 5,461.69
• 6.5 days @ $100.00/day vehicle and gas	$ 650.00
• 6.5 days @ $90.00/day room and board	$ 585.00
Total	**$ 6,696.69**

Total Trenching Costs	**$ 94,944.42**

(B) Technical

Line cutting

Sherwood Forest Contracting Services	
• 28 days @ $200.00/day wages	$ 5,600.00
• 28 @ $25.00/day power saw rental	$ 700.00
• 28 days @ $90.00/day room & board	$ 2,520.00
• 14 days @ $100.00/day vehicle and gas	$ 1,400.00
Total	**$ 10,220.00**
Brad Dondale	
• 4 days @ $200.00/day wages	$ 800.00
• 4 days @ $100.00/ day vehicle and gas	$ 400.00
• 4 @ $25.00/day power saw rental	$ 100.00
• 4 days @ $90.00/day room and board	$ 360.00
Total	**$ 1,660.00**
Total Line cutting costs	**$ 11,880.00**

Prospecting

Aaron Doyle - Prospector	
• 26 days @ $200.00/day wages	$ 5,200.00
• 26 days @ $90.00/day room & board	$ 2,340.00
• 26 days @ $100.00/vehicle & gas	$ 2,600.00
	$10,140.00
Jim Doyle – Prospector	
• 26 days @ $200.00/day wages	$ 5,200.00
• 26 days @ $90.00/day room & board	$ 2,340.00
	$ 7,540.00
Louis Doyle – Prospector	
• 26 days @ $300.00/day wages	$ 7,800.00
• 26 days @ 90.00/day room & board	$ 2,340.00
• 26 days @ $100.00/day vehicle & gas	$ 2,600.00
Total	**$12,740.00**
Total Prospecting costs	**$30,420.00**

Total Expenditures for (B) Technical **$137,244.42**

(C) Technical

Geological

Tim Barrett - Geologist

• 23 days @ $642.04/day wages	$14,767.00
• 17 days @ $90.00/day room & board	$ 1,530.00
• 12 days @ $100.00/day vehicle & gas	$ 1,200.00
Total	**$17,497.00**

Bruce Perry – Geologist

• 25 days @ $548.61/day wages	$13,715.25
• 10 days @ $90.00/day room and board	$ 900.00
• 10 days @ $100.00/day vehicle and gas	$ 1,000.00
Total	**$15,615.25**

Strathcona Minerals – Geologist

• 6.1 days @ $1,267.04/day wages	$ 7,728.94
• 6 days @ $90.00/day room and board	$ 540.00
• 6 days @ $100.00/day vehicle and gas	$ 600.00
Total	**$ 8,868.94**

Chris Wild - Wildrock Resources - Geologist

• 66 days @ $366.58/day wages	$24,194.28
• 59.5 days @ $100.00/day vehicle & gas	$ 5,950.00
• 66 days@ $90.00/day room and board	$ 5,940.00
Total	**$36,084.28**

Louis Doyle – Planning & Managing

• 53days @ $300.00/day wages	$15,900.00
• 53 days @ $90.00/day room & board	$ 4,770.00
• 43 days @ $100.00/day vehicle and gas	$ 4,300.00
Total	**$24,970.00**

Aaron Doyle – Mapping

• 21 days @ $280.00/day wages	$ 5,880.00
• 15 days @ $90.00/day room and board	$ 1,350.00
Total	**$ 7,230.00**

Aaron Doyle – Camp logistics

• 43 days @ $250.00/day wages	$10,750.00
• 43 days @ $90.00/day room & board	$ 3,870.00
• 31 days @ $100.00/day vehicle and gas	$ 3,100.00
Total	**$17,720.00**

(C) Technical

Jim Doyle – Camp logistics	
• 25 days @ $250.00/day wages	$ 6,250.00
• 25 days @ $90.00/day room & board	$ 2,250.00
Total	**$ 8,500.00**
Total Geological cost	**$136,485.47**

Geophysical

Peter Walcott & Associates – Geophysics	
Invoice #	
4306	$ 5,136.00
4305	$ 39,547.20
4287	$ 386.38
4286	$ 5,476.26
4290	$ 11,146.63
4335	$ 996.09
4336	$ 4,173.00
4337	$ 6,177.11
4338	$ 6,842.65
Total	**$ 79,881.32**
• 182 days @ $438.91/day wages	$ 79,881.62
• 134 days @ $90.00/day room and board	$ 12,060.00
• 25 days @ $100.00/day vehicle and gas	$ 2,500.00
Total	**$ 94,441.62**
Total Geophysical costs	**$ 94,441.62**

Drilling

Beaupre Diamond Drilling	
• 84 days @ $734.73/day wages	$ 61,717.32
• 84 days @ $90.00/day room and board	$ 7,560.00
• 42 days @ $100.00/day vehicle and gas	$ 4,200.00
Total	**$ 73,477.32**
Jack Heinemann	
• 23 days @ $262.61/day wages	$ 6,040.03
• 23 days @ $90.00/day room and board	$ 2,070.00
• 23 days @ $100.00/day vehicle and gas	$ 2,300.00
Total	**$ 10,410.03**
KCD Contracting – Invoiced	**$ 1,067.33**

(C) Technical

Rudy Geddert	
• 8 days @ $416.38/day wages	$ 3,331.05
• 8 days @ $90.00/day room and board	$ 720.00
• 8 days @ $100.00/day vehicle and gas	$ 800.00
Total	**$ 4,851.05**
Total Drilling costs	**$ 89,805.73**

Geochemical

Acme Analytical Laboritories	
• 154 assays @ $49.22/assay	**$ 7,580.43**
Eco-Tech Laboritories	
• 154 assays @ $29.45/assay	**$ 4,534.67**
ALS Chemex	
• 248 assays @ $68.49/assay	**$ 16,986.09**
Total Geochemical costs	**$ 29,101.19**

General Expenses

• Glynis Cox	$ 939.63
• Vehicle repair	$ 16,852.03
• Supplies	$ 5,044.20
• Groceries	$ 14,612.03
• Guard Dog	$ 875.86
• Travel Costs	$ 17,642.52
• Utilities	$ 5,779.56
Total	**$ 61,745.83**

Total expenditures for (C) Technical	**$411,579.84**
Total expenditures for Technical (B)	**$137,244.42**
Total expenditures for Technical (C)	**$411,579.84**
Total expenditures	**$548,824.26**

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716
(ISO 9002 Accredited Co.)

GEOCHEMICAL ANALYSIS CERTIFICATE

Barker Minerals Limited File # A200562
22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppb	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppb	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %
G-1	1.24	1.88	2.21	37.8	15	4.7	4.1	541	1.85	.3	1.5	.2	4.0	70.7	.01	.02	.40	41	.51
001-49 -80+230	2.21	44.05	24.27	169.8	341	104.3	31.1	902	2.80	27.6	1.4	1.4	11.3	38.4	.85	.28	45	7	.21
001-49 -230	3.21	51.99	34.61	214.1	283	118.3	28.7	1188	3.76	33.9	1.8	1.9	15.9	40.1	1.14	.38	.88	10	.21

SAMPLE#	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Sc ppm	Tl ppm	S %	Hg ppb	Se ppm	Te ppm	Ga ppm	Au** ppb	Pt** ppb	Pd** ppb
G-1	.087	7.7	46.1	.52	216.9	.119	<1	.89	.058	.45	.3	1.4	.28	.02	<5	.1	<.02	4.4	<2	3	3
001-49 -80+230	.073	21.7	9.0	.11	187.6	.001	<1	.29	.002	.05	<.1	1.8	.07	.17	31	1.6	03	.7	4	<2	<2
001-49 -230	.081	31.5	13.4	.17	213.9	.002	<1	.45	.002	.07	.1	2.7	.11	.07	50	2.1	.06	1.1	9	<2	2

GROUP 1F1 - 1.00 GM SAMPLE LEACHED WITH 6 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 20 ML, ANALYSED BY ICP/ES & MS.
UPPER LIMITS - AG, AU, HG, W, SE, TE, TL, GA, SN = 100 PPM; MO, CO, CD, SB, BI, TH, U, B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
AU** PT** & PD** BY FIRE ASSAY & ANALYSIS BY ICP-ES. (30 gm)
- SAMPLE TYPE: CLAY

DATE RECEIVED: MAR 5 2002 **DATE REPORT MAILED:** March 15/02 **SIGNED BY** C.L.........D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data ✓ FA ____

ACME ANA‴TICAL LABORATORIES LTD. 852 E. HASTINGS ST. V‴COUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604‵‶53-1716
(IS( )02 Accredited Co.)

GEOCHEMICAL AN(xSIS CERTIFICATE

Barker Minerals Limited File # A200563

22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppb	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppb	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm
G-1	1.72	1.87	2.70	36.0	13	4.5	3.8	562	1.70	.3	1.9	.2	5.1	87.2	.01	.02	.12	39
SS-01 -80+230	.64	30.91	24.69	65.7	120	34.3	11.0	396	2.07	15.5	1.0	4.1	5.8	15.2	.28	.24	.30	14
SS-01 -230	97	47.95	35.43	98.8	222	49.8	16.3	808	2.73	22.1	1.7	2.8	5 6	30.0	.58	.33	.43	21
SS-02 -80+230	.31	11.63	4.89	41.9	26	27.0	10.0	295	2.81	1.0	.4	.8	5.2	32.6	07	.02	12.13	40
SS-02 -230	.69	23 19	14.93	88.1	39	46 5	18 1	659	3.75	1.4	1.2	.8	30.8	73.1	.15	.05	50.20	57
SS-03 -80+230	.39	11.64	3.88	29.3	29	19.9	11.0	284	2.97	1 0	2.8	4	5.4	28 6	.06	.03	.91	40
SS-03 -230	.79	20.01	9.74	50.6	41	29.6	13.8	762	4.19	1 0	1.1	1.3	24 2	89 6	.10	.06	2.15	49
SS-04 -80+230	.33	11.36	6.16	34.9	39	17.6	8 6	271	2.62	1.6	.6	.3	6.7	30.8	09	.05	7.74	35
SS-04 -230	.85	24.91	14.42	71.4	75	29.2	13.7	559	3.13	3.6	1.1	1.6	14.4	67.0	.20	.10	23.53	36
SS-05 -80+230	.43	23.13	13.06	40.0	68	25.6	10.1	606	2.17	8.7	1.0	1.1	10.5	8.8	.08	.17	3.97	11
SS-05 -230	1.02	41.89	25.13	69.0	119	50.7	19.7	1499	3.79	13.2	2.0	3.9	12.4	15.8	.17	.27	2.29	18
SS-06 -80+230	1.20	48.18	25.86	88.1	211	37.3	14.2	424	2.86	23.9	1.5	1.1	6.6	18.5	.33	.28	12.38	19
SS-06 -230	1.83	63.52	38.39	127.0	271	49.7	18.6	712	3.94	32.3	2.2	3.7	10.9	27.6	.53	.31	7.65	30
RE SS-05 -80+230	.50	24.50	14.09	43.1	72	27.1	10.9	639	2.21	7 6	1.1	1.1	11.6	9 7	.09	.19	4.31	10
SS-07 -80+230	1.50	34.17	31.89	66.7	147	41.3	20.4	861	2.38	15.0	1.1	1.5	6.9	12.3	.45	.29	.82	17
SS-07 -230	2.67	62.04	54.76	122.0	217	73.6	37.0	1770	4.03	20.8	1.8	4.6	9.4	21.3	.65	.41	1.09	34
SS-08 -80+230	1.26	28.00	21.47	67.5	126	32.4	12.4	996	2.15	11.6	1.0	1.1	5.4	16.6	.33	.26	.42	15
SS-08 -230	1 89	40.71	28.76	95.7	233	46.6	18.3	2370	3.27	16.5	1.5	3.4	6.6	28.6	.79	.33	.51	24
SS-09 -80+230	1 06	42.52	36.05	95.6	176	43.0	15.0	460	2.75	21.3	1.3	1 6	5 5	16 4	.37	.31	.79	21
SS-09 -230	1.78	86.94	81.52	210.6	417	82.7	27.6	1084	5.03	40.8	2.6	2.7	7.0	26.9	83	.40	1.74	41
STANDARD DS3/FA-10R	9.09	122.28	34.60	150.1	298	35.2	12.1	818	3.19	29.4	5.9	20.0	3 8	26.9	5.18	5.02	5.61	75

SAMPLE#	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Sc ppm	Tl ppm	S %	Hg ppb	Se ppm	Te ppm	Ga ppm	Au** ppb	Pt** ppb	Pd** ppb
G-1	.63	.104	8.3	51.8	.54	213.7	.134	1	.99	.074	.44	.6	1.5	.28	.02	<5	<.1	<.02	4.4	<2	<2	<2
SS-01 -80+230	.29	.068	16.5	23.6	.40	39.7	.019	<1	.53	.002	.04	.3	1.0	.04	.06	10	.7	.02	1.6	80	<2	2
SS-01 -230	.58	.121	21.7	33.5	.53	74.5	.024	1	.81	.004	.07	.5	1.5	.06	.05	29	1.3	.03	2.3	154	11	4
SS-02 -80+230	.89	.152	12.3	35.5	.49	58.6	.063	1	.72	.015	.09	1.9	1.2	.07	.03	<5	.3	<.02	2.9	<2	<2	2
SS-02 -230	1.88	.244	63.6	62.6	.88	120.0	.117	3	1.43	.052	.18	4.4	2.9	.13	.04	11	<.1	<.02	5.5	<2	2	<2
SS-03 -80+230	.61	.081	11.6	22.1	.37	35.4	.042	<1	.59	.007	.07	.7	1.4	.06	.07	<5	.5	< 02	2.6	<2	<2	2
SS-03 -230	1.77	.094	44.3	33.4	.64	57.2	.080	<1	1.24	.020	.12	2.6	3.4	.10	.06	8	.5	<.02	4 0	<2	<2	4
SS-04 -80+230	1.06	.098	17.5	19.1	.60	35.8	.037	<1	.53	006	.06	.8	1.1	.05	.05	<5	.5	<.02	2.2	<2	6	<2
SS-04 -230	2.26	.145	34.7	29.6	1.17	74.3	.067	1	1.07	.016	.11	5.5	1.9	.09	.07	15	5	.02	3 6	<2	<2	5
SS-05 -80+230	.13	.038	27.5	12.5	.34	22.7	.017	<1	.63	.003	.04	.2	.7	.05	.02	10	.4	.03	1 9	2	<2	<2
SS-05 -230	.28	.093	37.1	21.8	.54	49.6	.022	<1	1.06	.004	.07	.6	1.5	.07	.02	13	.8	.06	3.0	38	4	3
SS-06 -80+230	.37	.117	16.7	24.4	.53	74.1	.022	<1	.73	.003	.05	.5	1.3	.04	.19	7	1.2	.02	2.3	<2	3	2
SS-06 -230	.54	.152	27.4	39.6	.81	132.4	.029	1	1.22	.006	.09	3.3	2.0	.07	.12	17	1.1	.03	3.6	3	<2	4
RE SS-05 -80+230	.13	.041	30.6	13.0	.35	26.1	.023	<1	.64	.005	.05	.6	.8	.04	.02	7	.5	.03	1.9	115	<2	<2
SS-07 -80+230	.21	.086	20.1	25.0	.36	60.8	.023	<1	.69	.003	.06	<.1	1.3	.06	.05	8	.8	.03	2 1	976	4	<2
SS-07 -230	.36	.121	31.0	46.3	.73	127.8	.036	<1	1.55	.004	.12	.2	2.4	.13	.03	23	.8	.04	4 5	5	<2	5
SS-08 -80+230	.28	.106	16.1	19.9	.34	72.0	.019	<1	.58	.002	.05	<.1	1.0	.04	.03	11	9	.02	1.8	2	5	<2
SS-08 -230	.45	.129	24.1	29.9	.53	131.6	.028	1	.98	.004	.08	<.1	1.8	.08	.03	30	1.2	.04	2.9	42	<2	<2
SS-09 -80+230	.34	.122	16.4	28.7	.52	67.9	.024	<1	.77	.002	.05	<.1	1.5	.05	.10	7	1.0	.03	2.3	<2	<2	<2
SS-09 -230	.58	.169	28.2	60.6	1.09	157.2	.038	<1	1.73	.003	.11	.3	3.6	.12	07	31	1 1	.05	5.0	6	<2	2
STANDARD DS3/FA-10R	.54	.092	17.1	180.2	.60	151.6	.082	2	1.74	.029	.16	3.5	2.6	1.11	.03	233	1.2	1.00	6.0	489	474	477

GROUP 1F1 - 1.00 GM SAMPLE LEACHED WITH 6 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 20 ML, ANALYSED BY ICP/ES & MS.
UPPER LIMITS - AG, AU, HG, W, SE, TE, TL, GA, SN = 100 PPM; MO, CO, CD, SB, BI, TH, U, B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
AU** PT** & PD** GROUP 3B BY FIRE ASSAY & ANALYSIS BY ICP-ES. (30 gm)
- SAMPLE TYPE: SEDIMENT Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: MAR 5 2002 DATE REPORT MAILED: March 15/02 SIGNED BY C. L...........D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data FA

ACME ANA TICAL LABORATORIES LTD. 852 E. HASTINGS ST. V COUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604 53-1716
(IS 002 Accredited Co.)

GEOCHEMICAL AN YSIS CERTIFICATE

Barker Minerals Limited File # A200564
22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppb	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppb	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm
SI	.10	1.18	4.93	4.0	25	.8	.1	6	.04	<.1	.1	73.5	<.1	1.7	.05	.27	.05	<2
SS-10	.57	9.62	9.19	44.8	25	35.9	15.2	621	8.81	4.2	2.0	14.1	16.3	24.5	.12	.84	.15	347
STANDARD DS3	9.40	123.32	33.79	151.6	273	35.4	11.2	819	3.24	29.1	6.0	19.0	4.1	27.7	5.72	4.86	5.62	77

SAMPLE#	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Sc ppm	Tl ppm	S %	Hg ppb	Se ppm	Te ppm	Ga ppm
SI	.76	.001	<.5	2.0	<.01	2.6	.001	<1	.01	.194	<.01	.2	<.1	<.02	.01	40	<.1	.03	.1
SS-10	.47	.068	11.1	198.0	.36	34.7	.161	3	.52	.017	.03	.2	2.1	.02	<.01	33	.2	<.02	4.5
STANDARD DS3	.53	.094	17.2	181.2	.58	146.3	.082	1	1.73	.028	.16	3.8	2.7	1.09	.02	206	1.4	1.02	6.1

GROUP 1F1 - 1.00 GM SAMPLE LEACHED WITH 6 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 20 ML, ANALYSED BY ICP/ES & MS.
UPPER LIMITS - AG, AU, HG, W, SE, TE, TL, GA, SN = 100 PPM; MO, CO, CD, SB, BI, TH, U, B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
- SAMPLE TYPE: BLACK SAND

DATE RECEIVED: MAR 5 2002 **DATE REPORT MAILED:** March 18/02 **SIGNED BY** C. L..........D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data ✓ FA ____

ACME ANA[illegible]TICAL LABORATORIES LTD. 852 E. HASTINGS ST. V[illegible]COUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604[illegible]53-1716
(IS[illegible]02 Accredited Co.)

ASSAY C[illegible]IFICATE

Barker Minerals Limited File # A200564
22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Ag** gm/mt	Au** gm/mt	Pt** gm/mt	Pd** gm/mt
SI	.3	.06	<.01	<.01
SS-10	<.3	9.17	<.01	<.01
STANDARD FA-10R	102.2	.47	.46	.46

GROUP 6 - PRECIOUS METALS BY FIRE ASSAY FROM 1 A.T. SAMPLE, ANALYSIS BY ICP-ES.
- SAMPLE TYPE: BLACK SAND

DATE RECEIVED: MAR 5 2002 DATE REPORT MAILED: March 19/02 SIGNED BY C. L[illegible] D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

 Data ✓ FA ___

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716
(ISO 9002 Accredited Co.)

WHOLE ROCK ICP ANALYSIS

Barker Minerals Limited File # A200561 Page 1

22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	SiO2 %	Al2O3 %	Fe2O3 %	MgO %	CaO %	Na2O %	K2O %	TiO2 %	P2O5 %	MnO %	Cr2O3 %	Ba ppm	Ni ppm	Sc ppm	LOI %	TOT/C %	TOT/S %	SUM %
001-01	46.75	10.03	15.77	5.76	14.41	2.18	1.03	1.97	.94	.26	.005	771	<20	23	.6	.04	.24	99.80
001-02	49.44	16.43	10.63	6.29	10.31	2.78	.21	.87	.19	.23	.050	96	90	37	2.4	.10	.02	99.86
001-03	51.77	16.63	6.87	2.13	5.41	3.89	5.36	.55	.41	.17	.004	2291	22	13	6.4	1.29	.19	99.85
001-04	47.61	15.81	10.02	6.43	8.09	3.25	3.31	.82	.30	.16	.017	750	50	26	4.0	.04	.04	99.91
001-05	46.86	13.50	10.71	6.03	7.17	3.17	3.29	.79	.44	.18	.020	995	39	28	7.6	1.37	.01	99.88
001-06	47.09	14.01	10.75	5.36	6.97	3.91	2.76	.77	.44	.18	.018	1122	21	24	7.5	1.62	.05	99.89
001-07	51.28	17.20	12.23	3.31	10.97	2.43	.11	1.63	.27	.19	.009	114	27	35	.2	.05	<.01	99.85
001-08	48.56	17.74	9.47	4.17	9.22	4.02	1.14	.95	.42	.16	.004	605	22	21	4.0	.06	.34	99.93
001-09	63.02	2.04	24.00	.06	.11	.28	.17	.17	.04	.01	.017	76	63	3	9.5	.07	11.83	99.44
001-10	45.44	3.41	30.82	1.79	.11	.21	.76	.05	.07	.18	.001	174	24	2	16.7	2.30	13.97	99.57
RE 001-10	44.76	3.38	31.70	1.76	.12	.19	.74	.05	.06	.18	<.001	174	24	2	16.6	2.36	13.97	99.56
001-11	47.76	11.44	23.90	3.03	.16	.10	1.35	.43	.12	.10	.010	944	60	9	11.4	.03	10.28	99.92
001-12	9.27	2.91	39.63	3.71	6.75	.03	.19	.15	.11	.96	.009	82	148	2	20.1	2.94	19.01	83.85
001-13	35.48	10.73	30.61	3.37	.51	1.15	.86	.47	.16	.19	.010	566	29	8	16.0	.37	14.74	99.61
001-14	29.19	10.25	17.01	7.06	12.46	.11	1.17	.41	.15	1.54	.019	443	54	8	14.7	4.54	1.94	94.13
001-15	70.11	10.49	6.49	1.92	2.44	1.51	1.51	1.09	.15	.16	.010	271	54	11	4.0	.82	.02	99.92
001-16	95.60	.47	2.80	.05	.10	.05	.14	.09	.08	.01	.015	108	<20	2	.6	.02	.37	100.02
001-17	84.90	.39	9.93	.10	.26	.05	.16	.05	.09	.03	.022	81	<20	1	3.9	.15	4.05	99.89
001-18	55.18	6.39	19.10	1.44	.37	.18	1.62	.56	.39	.03	.015	1194	87	7	14.3	.36	11.25	99.72
001-19	67.36	5.20	14.70	1.55	.72	.05	1.34	.42	.53	.02	.010	1385	89	5	7.6	.10	9.18	99.67
001-20	65.12	7.89	13.69	.85	.55	.06	2.98	.68	.48	.01	.024	3758	58	9	7.2	.04	8.06	99.96
001-21	67.09	2.53	17.98	2.02	.20	.02	.39	.34	.06	.07	.015	311	87	5	9.1	.20	8.84	99.86
001-22	93.32	.74	3.45	.23	.08	.10	.07	.07	.15	.01	.001	78	26	2	1.8	.02	.95	100.03
001-23	91.51	1.52	4.34	.38	.07	.05	.18	.12	.07	.01	.001	176	22	3	1.7	.01	.97	99.97
001-24	55.59	16.64	13.06	3.66	.25	3.39	1.20	.67	.09	.07	.022	1142	37	13	5.1	.01	2.70	99.88
001-25	12.05	3.50	39.88	3.66	5.53	.04	.34	.13	.06	.97	.012	153	37	3	20.3	2.98	18.33	86.49
001-26	62.85	3.28	21.49	.62	.41	.04	1.09	.27	.42	.01	.018	2221	127	4	8.8	.04	10.67	99.56
001-27	31.36	6.79	36.30	1.91	.06	.06	.88	.26	.12	.06	.006	599	30	5	21.8	.03	24.64	99.67
001-28	85.26	2.42	8.12	.15	.06	.07	.46	.21	.10	.02	.019	355	27	4	3.0	.05	1.86	99.94
001-29	88.06	.99	7.07	.63	.04	.02	.03	.06	.12	.01	.013	15	<20	3	2.9	.03	2.96	99.95
001-30	68.00	1.35	19.70	.38	.22	.03	.26	.08	.22	.01	.014	270	23	1	9.6	.12	12.13	99.90
STANDARD SO-17/CSB	61.51	14.00	5.88	2.33	4.66	4.15	1.32	.62	1.00	.56	.443	400	35	23	3.4	2.35	5.44	99.92

GROUP 4A - 0.200 GM SAMPLE BY LIBO2 FUSION, ANALYSIS BY ICP-ES. LOI BY LOSS ON IGNITION.
TOTAL C & S BY LECO. (NOT INCLUDED IN THE SUM)
- SAMPLE TYPE: ROCK R150 60C
Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: MAR 5 2002 **DATE REPORT MAILED:** March 22/02 **SIGNED BY** C. L. .... D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data FA

SAMPLE#	SiO2 %	Al2O3 %	Fe2O3 %	MgO %	CaO %	Na2O %	K2O %	TiO2 %	P2O5 %	MnO %	Cr2O3 %	Ba ppm	Ni ppm	Sc ppm	LOI %	TOT/C %	TOT/S %	SUM %
001-31	32.59	8.83	35.92	2.29	.62	1.32	.36	.31	.20	.07	.005	333	<20	5	16.7	.15	15.87	99.25
001-32	53.48	14.93	17.09	3.75	.20	1.57	1.32	.59	.03	.09	.008	1318	36	11	6.1	.03	3.60	99.32
001-33	59.53	3.99	24.47	1.32	.42	.04	.95	.36	.29	.02	.014	1431	<20	4	7.5	.02	13.30	99.07
001-34	66.16	7.00	11.21	3.69	.54	.20	1.38	.76	.23	.11	.072	1188	335	10	7.5	1.30	.23	99.03
001-35	13.12	.18	58.42	.02	.04	.05	.06	.01	.01	<.01	.002	32	<20	<1	27.1	.11	38.92	99.02
001-36	48.87	17.66	5.42	3.37	15.12	2.19	.34	.69	.10	.47	.011	76	83	14	4.8	1.74	.22	99.06
001-37	58.45	4.02	23.60	2.19	.55	.04	.72	.40	.29	.04	.005	743	<20	5	8.2	.08	12.83	98.59
001-38	38.48	<.03	38.42	.06	.06	.03	<.02	.01	<.01	.01	.006	10	<20	<1	22.3	.09	33.03	99.38
001-39	41.09	1.85	33.31	.13	.01	.03	.49	.11	<.01	<.01	.008	3651	<20	2	20.5	.02	28.98	97.94
001-39A	68.21	2.82	15.93	.22	.08	.03	.72	.17	.02	.01	.001	5368	<20	3	9.5	.03	12.06	98.32
001-40	65.27	3.81	18.54	.98	.28	.03	.97	.34	.17	.02	.005	1383	97	4	8.3	.04	10.93	98.88
RE 001-40	65.36	3.81	18.60	.98	.28	.03	.98	.34	.16	.02	.004	1384	110	4	8.2	.04	10.83	98.94
001-41	73.39	6.53	9.38	.43	.04	.06	1.70	.33	.06	<.01	.013	9844	22	6	5.7	.07	6.85	98.73
001-42	79.99	3.14	7.99	.23	<.01	.03	.79	.19	.03	<.01	.013	6181	33	3	4.9	.03	6.23	98.00
001-43	22.22	1.01	51.87	.08	.02	.01	.31	.07	<.01	<.01	<.001	1739	<20	1	23.8	.09	32.93	99.59
001-44	7.69	1.60	60.57	.77	.47	.13	.07	.17	.03	.06	.009	79	47	2	28.0	1.40	27.60	99.58
001-44A	29.10	2.19	43.33	1.00	1.68	.10	.43	.19	.07	.10	.023	619	<20	3	19.8	2.31	20.42	98.08
001-44B	48.04	.96	33.45	.82	1.87	.02	.27	.10	.06	.06	.021	260	<20	1	13.7	1.14	17.77	99.40
001-45	54.66	14.36	10.90	1.68	4.58	2.54	2.03	1.76	.29	.28	.007	941	38	18	6.3	1.65	.11	99.50
001-46	38.46	6.26	38.04	.99	.12	.13	.98	.58	.14	.07	.010	306	<20	6	12.6	.18	10.83	98.42
001-47	80.22	9.06	3.78	.35	.11	1.73	1.58	.39	.06	.04	.014	939	25	8	2.0	.11	.68	99.45
001-48	74.07	12.02	4.61	.57	.13	1.30	3.31	.61	.05	.08	.016	890	29	11	2.8	.18	.11	99.67
001-55	43.41	13.16	13.25	11.22	13.00	1.65	.55	1.55	.03	.18	.046	172	102	64	1.6	.02	.05	99.69
001-56	48.76	15.68	10.41	6.88	8.45	4.41	.92	.82	.29	.17	.022	408	40	29	2.7	.11	.55	99.57
001-57	41.76	11.50	15.17	13.12	11.76	1.59	.69	1.64	.62	.18	.037	208	68	40	1.4	.14	.04	99.50
001-58	47.12	17.05	12.78	4.16	12.50	1.63	.42	.62	.16	.29	.019	85	31	24	2.8	.05	2.04	99.57
001-59	44.48	14.39	12.72	8.90	12.32	2.04	.67	1.27	.53	.18	.028	246	45	38	2.0	.03	.07	99.56
001-60	46.78	16.01	12.87	5.69	13.23	1.54	.21	.66	.11	.33	.041	75	64	32	2.0	.05	.69	99.49
001-61	46.51	14.67	13.28	4.57	13.24	2.14	.43	2.33	.28	.15	.028	177	112	26	1.9	.02	.01	99.56
001-62	19.02	.31	53.01	.23	.18	.05	.05	.03	<.01	.01	.007	40	<20	<1	26.8	.27	39.93	99.71
001-63	1.75	.12	60.39	.10	.10	.03	<.02	.01	.02	.01	.004	12	<20	<1	30.0	.17	46.92	92.54
001-64	28.34	7.22	11.69	7.11	18.86	.64	1.01	1.04	.13	.24	.232	1028	836	17	22.7	7.55	.58	99.44
STANDARD SO-17/CSB	61.18	13.75	5.88	2.41	4.80	4.16	1.30	.61	.95	.55	.442	384	32	22	3.4	2.38	5.34	99.48

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

SAMPLE#	SiO2 %	Al2O3 %	Fe2O3 %	MgO %	CaO %	Na2O %	K2O %	TiO2 %	P2O5 %	MnO %	Cr2O3 %	Ba ppm	Ni ppm	Sc ppm	LOI %	TOT/C %	TOT/S %	SUM %
001-65	18.23	.43	52.84	.20	.10	.04	.13	.04	.07	.02	.008	60	<20	1	27.7	.21	42.87	99.82
001-66	64.99	9.59	11.87	.91	.21	.18	2.90	.32	.16	.14	.011	1082	99	12	7.6	1.00	6.11	99.01
001-67	6.94	<.03	58.21	<.01	.04	.01	<.02	.01	.01	<.01	.006	21420	<20	<1	27.4	.31	36.36	95.04
001-68	10.52	.50	57.92	.03	.03	.08	.12	.06	<.01	<.01	.003	2985	<20	1	28.8	.06	39.80	98.40
001-69	61.94	7.52	14.31	6.50	.20	.01	<.02	1.06	.11	.06	.188	47	502	20	7.3	.85	2.78	99.28
001-70	8.17	.51	58.41	.05	.06	.02	<.02	.06	.03	.01	.005	18271	<20	1	30.1	.03	46.41	99.48
001-71	72.45	3.33	12.60	.24	.04	.08	.88	.22	.04	.01	.013	5850	<20	4	7.6	.04	10.58	98.16
001-72	58.47	4.80	19.39	.30	.04	.06	1.21	.30	.04	.01	.007	7328	<20	5	12.7	.05	17.58	98.15
001-73	65.10	1.60	16.80	.06	.02	.47	.23	.15	.01	<.01	.020	14996	<20	2	11.7	.03	14.99	97.84
001-74	24.10	3.26	44.13	.23	.01	.04	.77	.21	.01	<.01	.006	6251	<20	3	24.5	.04	32.46	97.96
001-75	30.45	3.46	40.40	.25	<.01	.04	.81	.24	.01	<.01	.010	6728	<20	3	22.2	.04	29.95	98.63
001-76	13.76	2.57	53.36	.17	.04	.07	.70	.18	.03	<.01	.002	4273	<20	2	27.2	.03	37.43	98.56
RE 001-76	14.06	2.62	52.84	.17	.04	.07	.73	.18	.05	<.01	.003	4400	<20	2	27.5	.04	38.32	98.76
001-77	40.66	2.47	39.33	.48	.03	.01	.02	.09	.13	.07	.010	145	22	3	16.6	.14	19.48	99.92
001-78	26.33	1.01	9.42	7.69	13.72	.08	.18	.11	.06	.55	.029	477	114	3	15.0	7.44	3.02	74.24
001-79	12.12	1.05	56.07	.10	.08	.07	.26	.09	.06	.01	.004	2384	<20	1	28.5	.08	40.73	98.68
001-80	20.96	2.26	47.09	.15	.04	.03	.54	.13	.03	<.01	.003	4774	<20	1	26.0	.03	37.29	97.77
001-81	74.43	2.62	15.21	.05	.08	.37	.17	.22	.06	.01	.018	113	33	3	6.5	.08	7.06	99.76
001-82	11.48	3.12	55.17	.17	.03	.06	.71	.22	.04	.01	.003	635	<20	3	28.7	.19	43.62	99.79
001-83	14.04	2.98	53.35	.19	.03	.06	.69	.22	.07	.01	.003	614	<20	3	28.0	.20	42.22	99.71
001-84	11.87	.97	55.43	.21	.06	.04	.21	.05	.10	.01	.004	197	<20	1	28.8	.29	43.71	97.78
001-85	48.42	23.15	7.36	2.89	.18	.64	6.80	.87	.11	.04	.015	1435	56	19	9.1	1.97	.42	99.75
001-86	8.32	1.43	1.28	1.74	46.87	.17	.31	.06	.13	.05	<.001	204	<20	1	35.0	11.50	.18	95.39
001-87	18.69	5.44	50.07	1.24	.50	.36	.70	.29	.34	.08	.012	122	<20	5	18.1	.68	19.62	95.84
001-88	75.65	9.95	4.27	.55	.41	.12	2.80	.44	.08	.02	.013	1804	53	8	5.2	2.29	.97	99.71
001-89	91.64	2.64	.87	.18	.16	.05	.73	.11	.08	<.01	.015	770	<20	3	3.2	2.38	.06	99.76
001-90	51.41	12.09	16.40	4.08	4.03	.41	1.07	1.90	.32	.15	.030	1045	70	17	7.7	.06	3.70	99.71
001-91	64.99	17.03	3.91	1.21	.42	.16	4.80	.78	.12	.02	.015	2347	59	14	6.1	1.18	1.43	99.83
001-92	76.23	7.48	4.26	1.39	2.63	.07	.97	.35	.24	.03	.012	889	35	7	5.8	1.69	1.49	99.57
001-93	54.06	5.13	27.39	1.22	.18	.10	.50	.24	.11	.04	.006	151	33	5	10.4	.09	8.72	99.40
001-94	77.32	9.25	5.24	1.56	.25	.48	1.15	.36	.07	.01	.013	274	52	8	4.0	.32	2.63	99.74
001-95	55.40	16.81	11.18	1.36	2.22	4.55	1.28	.84	.10	.04	.014	259	33	10	5.9	.09	2.76	99.73
STANDARD SO-17/CSB	61.33	14.20	5.83	2.23	4.58	4.15	1.42	.59	1.03	.56	.429	417	36	25	3.4	2.38	5.32	99.80

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data ✓ FA ____

SAMPLE#	SiO2	Al2O3	Fe2O3	MgO	CaO	Na2O	K2O	TiO2	P2O5	MnO	Cr2O3	Ba	Ni	Sc	LOI	TOT/C	TOT/S	SUM
	%	%	%	%	%	%	%	%	%	%	%	ppm	ppm	ppm	%	%	%	%
001-86	7.83	1.44	1.21	1.72	48.36	.19	.32	.06	.13	.05	.001	201	<20	2	38.6	11.40	.12	99.93
001-87	21.77	6.50	44.52	1.44	.51	.41	.67	.36	.33	.10	.012	143	20	7	18.0	.68	18.53	94.64
001-88	76.47	9.83	3.92	.52	.34	.11	2.82	.45	.20	.02	.013	1870	52	9	5.0	2.22	.84	99.91
001-89	91.14	2.87	.92	.19	.17	.10	.72	.13	.11	<.01	.014	844	<20	3	3.3	2.46	.04	99.76
001-90	51.00	12.41	16.53	3.72	3.94	.41	1.13	2.02	.31	.15	.032	1090	65	19	7.9	.08	3.46	99.68
001-91	64.13	17.66	3.83	1.21	.44	.16	4.70	.84	.18	.03	.015	2587	50	15	6.4	1.23	1.39	99.89
001-92	75.83	7.93	4.35	1.38	2.63	.06	.99	.38	.33	.04	.012	962	36	8	5.7	1.70	1.32	99.75
001-93	53.45	5.30	28.09	1.17	.17	.12	.53	.26	.14	.04	.006	152	<20	5	10.2	.10	8.38	99.50
001-94	76.91	9.68	5.27	1.53	.24	.49	1.28	.41	.12	.01	.013	287	48	9	3.9	.36	2.50	99.89
001-95	54.86	17.20	11.27	1.30	2.14	4.51	1.38	.89	.16	.04	.014	268	37	11	6.0	.07	2.56	99.80
001-96	66.88	15.10	5.32	1.25	.89	6.48	.56	.70	.17	.03	.013	211	49	8	2.4	.07	1.34	99.82
RE 001-96	66.99	15.09	5.27	1.26	.90	6.57	.58	.72	.15	.03	.013	212	48	8	2.2	.07	1.28	99.80
001-97	60.61	16.02	8.41	2.64	.37	.65	4.49	.81	.12	.12	.012	760	115	16	5.6	.56	2.10	99.95
001-98	66.18	17.18	3.59	1.15	.31	.22	5.25	.66	.20	.03	.012	2758	31	11	4.9	.68	.60	100.00
001-99	71.07	10.56	7.31	.60	.48	.28	3.23	.51	.08	.02	.011	963	51	9	5.3	.35	4.85	99.57
001-100	36.18	9.25	10.03	6.92	15.05	.06	1.15	1.40	.23	.17	.207	454	394	27	19.0	4.72	.07	99.75
001-101	59.92	11.60	12.97	1.10	.64	.25	3.52	.59	.11	.03	.012	769	66	12	9.0	.59	8.97	99.84
001-102	84.48	5.28	2.84	.35	.77	.12	1.01	.23	.29	.01	.018	1125	76	6	3.6	2.27	<.01	99.13
001-103	12.96	.23	56.99	.04	.08	.03	.03	.02	.04	<.01	.003	1662	<20	<1	29.2	.01	43.70	99.81
001-104	77.19	9.59	3.23	.71	.44	.18	2.22	.49	.09	.01	.015	2590	48	9	5.3	2.81	1.25	99.76
001-105	40.93	.75	7.82	37.75	.17	.01	<.02	.01	.01	.07	.342	32	1843	7	11.7	.11	.01	99.81
001-106	55.86	16.79	8.50	4.41	2.68	3.96	1.70	1.02	.17	.17	.007	735	32	25	4.4	.10	.90	99.76
001-107	92.77	<.03	4.31	.08	.07	.05	.02	.02	.03	.01	.003	14	32	<1	2.5	.04	2.60	99.87
001-108	41.56	1.13	9.25	35.22	.99	<.01	<.02	.02	<.01	.06	.356	30	2055	7	11.0	.46	.03	99.85
001-109	42.37	.66	8.32	36.56	.02	<.01	<.02	.01	<.01	.07	.357	10	1738	8	11.3	.06	<.01	99.89
001-110	24.09	1.93	48.63	.14	.01	.32	.43	.13	<.01	<.01	.002	136	<20	1	24.2	.01	35.72	99.90
001-111	48.05	12.10	22.68	3.60	.18	.40	.84	1.40	.07	.41	.038	216	102	19	9.7	2.10	.84	99.51
001-112	26.11	11.76	33.40	7.27	.33	.29	.60	1.73	.16	.69	.057	150	109	24	16.7	3.52	3.58	99.13
001-113	63.71	12.02	11.48	2.45	.22	.38	1.20	1.39	.10	.13	.063	344	123	19	6.0	1.41	.22	99.20
001-114	79.44	4.68	9.03	.26	.04	.31	.73	.25	<.01	.04	.020	189	29	6	5.0	2.48	.61	99.83
001-115	26.90	6.94	41.84	1.31	.28	.38	.74	.38	.29	.04	.014	179	41	7	17.4	.55	19.86	96.54
001-116	24.76	4.20	47.70	.17	.04	.10	.90	.26	.05	.01	.016	402	<20	3	21.5	.20	28.88	99.75
STANDARD SO-17/CSB	61.39	13.94	5.75	2.30	4.63	4.14	1.45	.63	.92	.57	.424	404	36	24	3.4	2.47	5.35	99.60

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

SAMPLE#	SiO2 %	Al2O3 %	Fe2O3 %	MgO %	CaO %	Na2O %	K2O %	TiO2 %	P2O5 %	MnO %	Cr2O3 %	Ba ppm	Ni ppm	Sc ppm	LOI %	TOT/C %	TOT/S %	SUM %
001-117	65.07	15.87	5.80	1.28	.39	.39	4.81	.65	.08	.03	.016	1448	37	12	5.3	.61	2.51	99.86
001-118	22.20	.83	49.88	.17	.07	.04	.23	.06	.04	.01	.007	107	<20	1	26.2	.18	39.38	99.75
001-119	85.13	2.18	4.00	1.41	2.75	.04	.17	.16	.10	.05	.030	232	30	5	3.9	1.11	.02	99.95
001-120	42.42	14.37	13.01	3.69	6.45	.09	1.53	2.29	.21	.20	.084	1051	217	32	15.4	3.30	.19	99.89
001-121	69.74	9.84	6.91	.57	3.96	.09	.85	1.22	.14	.08	.044	475	54	15	6.4	1.14	.01	99.91
001-122	42.88	5.34	9.59	11.15	8.73	.04	.11	.83	.04	.19	.271	83	589	19	20.7	6.28	<.01	99.96
001-123	47.35	15.15	13.33	4.35	12.14	2.77	.45	2.12	.25	.15	.050	198	183	28	1.8	.04	<.01	99.96
001-124	39.04	.39	7.51	40.07	.14	.01	.02	.02	.04	.05	.347	5	1815	7	12.1	.31	.01	99.96
001-125	23.38	1.82	48.20	.30	.09	.05	.49	.11	.04	.01	.010	216	<20	2	25.3	.23	37.63	99.83
001-126	49.36	15.83	10.51	7.02	7.10	2.70	2.02	.82	.23	.16	.034	1151	60	28	4.0	.42	.41	99.92
RE 001-126	49.34	15.68	10.39	7.10	7.12	2.68	1.88	.80	.19	.16	.032	1128	59	27	4.2	.39	.42	99.71
STANDARD SO-17/CSB	61.46	13.95	5.85	2.33	4.64	4.12	1.41	.62	.99	.56	.443	399	34	23	3.4	2.45	5.46	99.83

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data d FA ____

ACME ANA TICAL LABORATORIES LTD. 852 E. HASTINGS ST. V COUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604 53-1716
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GEOCHEMICAL AN SIS CERTIFICATE

Barker Minerals Limited File # A200561 Page 1 (a)

22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Co ppm	Cs ppm	Ga ppm	Hf ppm	Nb ppm	Rb ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm	W ppm	Zr ppm	Y ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tm ppm	Yb ppm	Lu ppm
001-01	41.6	1.1	19.8	5.4	13.7	29.6	3	649.9	1.4	4.8	2.2	337	<1	143.4	70.9	47.5	116.1	15.75	76.3	17.1	4.55	15.36	2.03	12.46	2.26	6.54	.95	6.69	.93
001-02	39.6	.2	16.7	1.4	1.8	4.3	<1	357.0	.1	.4	.6	273	<1	54.6	22.1	6.2	12.1	1.81	9.5	2.8	.95	3.26	.53	3.92	.76	2.18	.32	2.20	.40
001-03	16.0	1.3	17.0	1.8	2.2	67.9	<1	789.0	.2	1.2	1.2	231	<1	60.6	19.4	9.7	18.4	2.37	10.9	3.0	.96	3.16	.44	3.30	.65	1.89	.30	1.96	.31
001-04	34.2	.9	15.6	1.7	1.8	45.4	<1	448.5	.1	.7	.6	244	<1	56.3	20.4	5.8	11.9	1.72	9.2	2.8	.95	3.08	.52	3.29	.73	1.98	.29	2.01	.33
001-05	32.4	4.0	14.2	1.7	1.5	57.4	<1	507.0	.1	.7	.3	258	<1	47.0	20.2	7.2	14.7	1.98	10.9	2.8	1.00	3.56	.51	3.37	.65	1.90	.27	2.05	.31
001-06	30.7	3.2	14.8	1.7	1.7	44.5	<1	689.3	.1	.8	.6	241	<1	47.4	19.9	7.0	13.6	1.98	10.3	2.8	.92	3.21	.51	3.29	.66	1.90	.28	1.84	.28
001-07	27.7	.2	22.6	3.2	2.5	2.9	<1	219.4	.2	.4	.5	330	<1	114.3	43.1	6.1	15.4	2.56	15.3	5.3	1.71	6.20	1.01	7.01	1.45	4.32	.63	4.20	.66
001-08	22.0	.3	18.5	2.1	2.7	19.5	<1	203.0	.2	1.0	1.3	298	<1	78.2	23.0	9.1	19.0	2.65	13.3	3.9	1.18	3.79	.54	4.08	.79	2.32	.32	2.23	.35
001-09	19.5	.4	3.7	3.0	2.1	7.4	2	23.4	.1	4.3	.8	15	<1	108.1	6.3	14.7	28.4	3.15	12.0	2.2	.30	1.66	.22	1.10	.20	.54	.09	.60	.09
001-10	81.2	.7	3.4	1.1	.6	32.2	8	18.9	<.1	2.9	2.3	<5	3.3	37.8	7.8	8.2	18.1	1.83	7.8	1.4	.19	1.46	.18	1.24	.25	.67	.09	.52	.10
RE 001-10	78.9	.7	3.1	1.0	.6	27.5	6	18.0	<.1	2.5	2.6	<5	3.1	36.4	9.2	7.7	16.1	1.79	6.5	1.5	.22	1.63	.24	1.52	.30	.80	.11	.63	.09
001-11	79.6	2.5	16.2	1.5	8.2	72.4	3	16.2	.6	7.9	1.5	50	1.4	57.0	14.0	29.4	49.2	5.78	23.6	4.3	1.04	3.41	.44	2.53	.45	1.30	.17	1.28	.19
001-12	92.7	.4	4.5	.7	2.2	10.3	3	100.0	.2	1.8	.9	20	1.7	25.5	10.9	6.3	12.1	1.47	6.4	1.4	.99	1.65	.25	1.67	.31	.86	.13	.85	.10
001-13	62.4	1.6	15.7	2.3	8.8	43.1	3	52.4	.7	8.9	2.3	60	1.5	79.5	17.5	31.8	56.5	6.56	25.9	4.6	1.21	3.93	.48	2.81	.53	1.46	.21	1.36	.21
001-14	24.8	1.3	14.7	2.2	6.1	55.4	7	167.0	.5	6.7	1.7	71	1.8	81.3	28.4	19.4	36.0	4.34	18.2	4.1	1.91	4.57	.68	4.36	.90	2.28	.34	2.18	.35
001-15	15.6	1.3	14.9	7.3	14.1	73.2	1	101.1	1.3	11.1	2.2	68	2.6	248.4	24.7	32.3	63.8	7.35	32.0	6.2	1.55	5.60	.79	4.52	.82	2.02	.27	1.92	.30
001-16	8.3	.2	1.7	2.4	1.8	5.8	2	7.2	.1	1.5	.5	8	2.4	85.8	7.8	8.9	21.1	2.02	8.6	1.5	.26	1.75	.25	1.36	.24	.68	.09	.62	.09
001-17	61.5	.2	1.2	.9	1.1	7.1	2	5.8	<.1	.9	.7	8	1.8	30.3	4.7	5.3	10.8	1.10	4.5	1.1	.20	1.07	.16	.87	.14	.34	.06	.37	.06
001-18	22.8	1.2	10.6	2.3	10.9	50.1	7	22.6	.8	5.5	5.2	202	2.8	77.3	14.2	11.1	19.4	2.54	10.6	2.0	.57	1.95	.29	1.85	.43	1.19	.19	1.29	.19
001-19	13.3	.6	9.2	1.4	8.2	32.5	10	13.0	.6	3.8	4.0	189	5.6	55.2	13.5	11.4	20.0	2.69	10.6	2.4	.50	2.50	.37	1.91	.42	1.03	.14	1.04	.17
001-20	14.3	1.0	13.0	2.2	13.8	56.5	16	17.5	1.1	4.6	5.3	150	3.3	78.0	15.2	21.3	35.3	4.46	17.4	3.4	.74	2.92	.41	2.40	.47	1.33	.18	1.21	.20
001-21	37.4	.3	5.1	.6	3.6	9.3	4	4.3	.2	.9	1.4	76	2.0	22.8	8.9	7.3	12.2	1.58	7.2	1.7	.46	1.92	.26	1.44	.27	.75	.09	.56	.09
001-22	11.6	.3	3.1	1.5	1.1	3.5	2	8.1	<.1	1.4	.4	5	5.2	53.4	8.6	7.9	18.2	1.79	7.3	1.5	.27	1.60	.25	1.44	.27	.73	.11	.67	.12
001-23	8.9	.3	4.2	1.5	2.4	7.7	1	6.2	.2	1.7	.7	11	4.4	50.4	5.0	14.4	31.2	3.10	12.6	2.4	.37	1.58	.19	1.06	.16	.36	.06	.49	.08
001-24	12.1	2.0	22.3	2.9	13.5	48.5	3	112.7	1.0	10.0	1.9	64	2.0	96.8	16.2	25.9	45.5	5.38	21.0	3.6	1.08	2.74	.43	2.45	.53	1.51	.24	1.56	.24
001-25	278.5	.8	5.1	.8	2.3	22.0	4	97.4	.2	3.1	1.9	27	2.7	29.4	20.2	13.5	24.2	2.88	12.2	2.8	1.08	2.84	.49	3.00	.65	1.62	.24	1.49	.25
001-26	19.6	.6	6.7	1.1	4.6	21.4	12	9.5	.3	2.6	4.3	144	2.4	35.4	9.5	12.6	22.5	2.76	11.1	2.1	.52	1.94	.28	1.44	.28	.77	.12	.69	.12
001-27	117.9	1.5	9.6	1.0	4.7	44.2	2	10.5	.4	5.3	.9	31	2.1	33.9	10.4	19.6	31.9	3.75	14.2	2.6	.69	2.00	.33	1.81	.32	.90	.14	.91	.13
001-28	23.9	.7	5.4	2.6	4.6	16.6	4	14.8	.3	4.0	1.0	24	4.0	91.3	10.4	19.2	42.5	4.36	18.1	3.6	.59	2.60	.42	2.05	.37	.92	.15	1.04	.16
001-29	42.0	<.1	4.4	1.1	1.7	1.0	2	2.9	<.1	1.9	.4	14	1.6	44.4	10.6	14.0	32.4	3.03	12.0	2.2	.39	2.00	.36	1.85	.35	.92	.12	.92	.15
001-30	159.6	.3	7.2	.7	1.8	7.6	11	14.7	.1	1.3	3.4	159	4.6	24.9	3.2	5.3	9.7	1.07	4.1	.7	.16	.69	.08	.50	.08	.27	<.05	.40	.06
STANDARD SO-17	19.4	3.9	19.4	11.7	24.8	22.7	11	323.9	4.5	11.1	12.7	124	11.1	365.2	27.4	10.9	23.2	2.87	13.8	3.3	1.07	3.80	.64	4.34	.96	2.78	.41	2.89	.46

GROUP 4B - REE - LiBO2 FUSION, ICP/MS FINISHED.
- SAMPLE TYPE: ROCK R150 60C
<u>Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.</u>

DATE RECEIVED: MAR 5 2002 **DATE REPORT MAILED:** March 22/02 **SIGNED BY** C. L. ....D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Data L FA ____

SAMPLE#	Co ppm	Cs ppm	Ga ppm	Hf ppm	Nb ppm	Rb ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm	W ppm	Zr ppm	Y ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tm ppm	Yb ppm	Lu ppm
001-31	37.9	.7	12.1	1.2	5.9	14.4	<1	54.0	.4	5.6	1.4	26	<1	43.3	12.9	19.0	40.0	4.19	17.9	3.5	.99	2.66	.36	2.18	.39	1.10	.15	.93	.14
001-32	17.4	2.4	22.5	2.5	11.5	57.0	5	60.3	.9	12.6	2.5	52	<1	84.9	16.4	45.6	80.5	8.18	31.2	5.9	1.62	4.10	.56	3.01	.55	1.40	.22	1.54	.22
001-33	4.6	.4	7.2	.9	6.6	19.5	8	7.8	.4	3.3	3.2	115	1.8	39.6	10.7	12.2	20.3	2.47	10.8	2.2	.29	1.97	.26	1.62	.29	.86	.11	.72	.10
001-34	41.5	1.7	11.2	2.3	14.0	60.0	9	41.2	.8	4.5	6.3	98	4.1	77.9	14.7	19.1	30.5	3.89	15.7	3.4	.81	2.57	.37	2.59	.44	1.30	.20	1.45	.21
001-35	6.7	<.1	<.5	<.5	<.5	2.5	52	2.7	<.1	<.1	.5	<5	1.1	2.2	.5	<.5	<.5	.07	<.4	<.1	<.05	.15	.01	.10	<.05	<.05	<.05	<.05	<.01
001-36	27.0	.8	23.0	2.8	13.1	26.2	12	611.9	.9	13.6	10.1	62	<1	91.3	31.7	57.4	102.9	11.52	46.3	9.1	1.57	6.92	1.04	5.73	1.00	2.69	.37	2.58	.38
001-37	11.2	.3	6.8	1.1	6.7	20.2	5	8.5	.4	2.2	3.8	99	3.5	39.1	10.7	12.5	20.4	2.45	10.1	2.3	.48	1.86	.25	1.82	.31	.83	.10	.70	.10
001-38	101.6	<.1	<.5	<.5	<.5	.5	<1	.8	<.1	<.1	.2	<5	<1	.5	<.1	<.5	<.5	<.02	<.4	<.1	<.05	<.05	<.01	<.05	<.05	<.05	<.05	<.05	<.01
001-39	3.5	.6	4.8	.7	1.8	21.9	286	8.3	.1	2.1	2.3	168	12.1	25.6	5.5	8.2	13.7	1.76	6.8	1.2	.24	.88	.12	.83	.16	.43	.06	.38	.06
001-39A	8.9	.8	6.2	1.1	2.7	35.0	187	9.6	.2	3.1	2.7	251	14.2	39.0	6.5	11.9	20.4	2.45	9.6	1.8	.40	1.22	.15	1.06	.21	.56	.08	.53	.07
001-40	17.4	.4	7.1	1.1	6.1	22.4	14	6.8	.4	2.7	2.6	131	4.3	37.9	6.3	7.5	11.8	1.66	7.0	1.4	.22	1.22	.16	1.06	.19	.53	.07	.52	.07
RE 001-40	17.8	.5	6.5	1.0	6.1	22.0	13	7.0	.4	2.5	3.0	132	4.5	37.7	6.5	7.9	11.6	1.79	7.1	1.4	.20	1.24	.17	1.06	.19	.57	.06	.46	.08
001-41	9.2	1.7	12.7	4.2	6.0	76.1	123	20.5	.5	9.1	3.4	92	10.7	145.0	14.1	26.7	52.3	5.20	20.5	3.8	.46	2.38	.35	2.17	.39	1.19	.17	1.10	.19
001-42	6.8	1.0	6.7	1.1	2.7	39.0	216	7.1	.2	2.8	3.3	278	11.9	36.5	6.5	12.7	18.6	2.27	8.9	1.8	.40	1.24	.17	1.15	.18	.54	.06	.59	.08
001-43	25.5	.4	2.6	.5	.8	12.0	129	6.2	<.1	1.0	.9	46	11.8	16.9	2.8	3.7	4.5	.80	3.4	.6	.17	.56	.07	.50	.07	.22	<.05	.16	.02
001-44	93.0	.3	4.7	<.5	2.5	3.3	6	36.3	.1	.4	.9	12	11.8	12.8	11.6	16.3	23.5	2.72	10.8	1.8	.32	1.77	.26	1.75	.27	.82	.09	.54	.05
001-44A	116.9	.5	6.0	.6	2.5	17.1	7	34.3	.2	.9	2.0	28	7.9	20.1	11.6	5.9	6.6	1.38	5.6	1.4	.15	1.33	.22	1.40	.29	.82	.08	.65	.08
001-44B	23.9	.3	3.3	<.5	1.2	12.3	7	34.0	<.1	.4	1.2	15	3.8	9.4	4.8	1.3	<.5	.48	2.0	.4	.08	.67	.07	.61	.12	.33	<.05	.20	.04
001-45	22.0	2.6	18.8	6.3	28.4	84.1	2	204.9	1.9	10.5	2.9	145	5.4	236.3	34.1	38.5	85.8	9.18	37.1	8.1	1.56	7.27	1.05	6.61	1.06	2.90	.41	3.00	.42
001-46	94.5	.9	9.9	2.7	16.8	46.1	96	33.9	1.0	6.5	1.6	29	6.7	98.8	18.2	35.7	74.1	7.60	30.8	6.0	.60	4.84	.65	3.68	.54	1.58	.21	1.44	.19
001-47	7.0	1.3	10.7	6.3	5.3	67.9	1	46.2	.6	12.3	2.9	27	<1	230.7	16.7	32.7	70.0	6.91	25.4	4.5	.86	3.33	.50	2.71	.46	1.41	.21	1.56	.23
001-48	10.7	2.0	15.7	9.0	9.9	114.6	<1	39.5	.9	15.6	3.3	50	<1	317.6	20.1	37.1	77.1	7.89	29.1	5.3	.88	4.28	.61	3.61	.61	1.82	.26	1.92	.31
001-55	52.5	.1	14.5	1.3	1.4	3.4	<1	328.2	<.1	<.1	<.1	453	<1	28.6	28.3	3.5	10.5	2.21	12.9	4.5	1.27	5.02	.80	5.25	.93	2.60	.34	2.28	.39
001-56	34.3	2.1	16.3	1.5	3.2	29.2	<1	537.2	.2	1.9	1.4	238	<1	45.3	18.1	10.8	20.6	2.80	11.8	3.2	.99	3.43	.54	3.16	.59	1.51	.24	1.59	.23
001-57	53.3	<.1	15.7	2.2	3.6	4.0	1	453.0	.2	.3	.1	446	<1	47.7	39.6	8.1	25.7	4.56	26.1	8.1	2.02	8.35	1.31	7.50	1.30	3.58	.49	3.19	.47
001-58	26.4	.4	19.2	1.0	<.5	10.1	3	300.1	<.1	.6	1.2	250	1.0	32.1	18.5	4.1	7.4	1.43	7.1	2.1	.68	2.72	.42	2.98	.59	1.62	.23	1.66	.31
001-59	43.9	.1	17.5	1.5	2.1	7.6	1	808.8	<.1	.3	.1	380	<1	36.3	32.7	10.9	28.9	4.61	23.0	6.9	1.75	6.89	1.03	6.12	1.00	2.96	.38	2.61	.40
001-60	32.5	<.1	17.9	1.1	.5	3.0	2	372.2	<.1	.9	.9	285	1.0	34.5	18.1	6.6	12.4	2.01	9.5	3.0	.76	2.87	.44	3.02	.62	1.62	.24	1.65	.28
001-61	45.4	.5	22.6	4.0	33.0	10.4	4	806.3	2.1	3.6	1.0	267	<1	150.5	24.6	28.5	60.0	6.77	26.9	6.2	1.75	5.80	.88	5.01	.80	2.25	.29	1.91	.28
001-62	11.0	.2	<.5	<.5	<.5	2.4	130	4.5	<.1	.1	.7	<5	1.8	4.6	1.1	<.5	<.5	.12	.8	<.1	<.05	.20	.02	.21	<.05	.07	<.05	<.05	<.01
001-63	63.5	.1	.7	<.5	<.5	.7	205	1.8	<.1	<.1	1.8	<5	2.9	.5	1.8	<.5	<.5	.02	<.4	<.1	<.05	.22	.03	.32	<.05	.10	<.05	<.05	<.01
001-64	98.3	1.0	10.1	1.6	11.8	41.5	46	391.2	.6	1.3	4.3	118	7.3	57.7	20.9	10.7	21.3	2.76	11.4	2.8	.68	3.38	.58	3.31	.61	1.64	.21	1.41	.20
STANDARD SO-17	18.8	4.0	20.3	11.9	26.0	22.9	12	323.1	4.2	12.3	12.1	123	12.1	367.0	27.1	12.0	24.4	3.01	14.2	3.5	1.04	4.04	.62	4.43	.94	2.90	.40	3.00	.45

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

ACME ANALYTICAL

Barker Minerals Limited FILE # A200561

ACME ANALYTICAL

SAMPLE#	Co ppm	Cs ppm	Ga ppm	Hf ppm	Nb ppm	Rb ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm	W ppm	Zr ppm	Y ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tm ppm	Yb ppm	Lu ppm
001-65	20.3	.2	.8	<.5	.9	5.0	14	4.6	<.1	.6	2.5	11	4.2	8.6	1.8	1.5	3.9	.30	1.4	.4	.09	.56	.06	.38	.08	.16	<.05	.17	.03
001-66	37.3	2.5	11.3	6.4	6.4	98.9	21	30.5	.4	11.3	4.9	60	1.4	210.6	23.4	22.4	45.2	4.71	20.8	4.4	.82	5.16	.76	4.51	.78	2.13	.30	1.84	.27
001-67	<.5	<.1	4.9	<.5	<.5	.8	1000	77.2	<.1	<.1	2.6	19	9.1	3.4	.6	.6	.5	.04	<.4	.2	.98	.21	.01	.19	<.05	<.05	<.05	.06	.02
001-68	.9	.2	2.3	<.5	1.1	3.8	251	27.6	<.1	1.1	6.1	63	8.0	15.0	1.4	3.0	5.1	.50	2.1	.2	.16	.34	.04	.23	.05	.12	<.05	.16	.03
001-69	63.4	.1	11.7	1.8	10.5	1.5	25	3.4	.5	1.0	.9	137	11.9	58.8	13.1	10.8	23.0	2.59	12.3	2.5	.48	2.34	.40	2.41	.50	1.31	.15	1.17	.16
001-70	3.8	.5	3.8	.6	1.1	2.1	495	111.1	.1	1.2	3.9	105	43.8	14.8	2.5	7.0	10.8	1.25	4.8	.7	1.64	.75	.08	.51	.08	.20	<.05	.26	.03
001-71	1.8	1.1	7.0	1.3	3.8	37.5	75	6.6	.3	3.6	2.5	102	15.0	43.3	4.9	11.2	21.1	2.29	9.4	1.4	.31	1.10	.15	.77	.16	.49	.05	.48	.07
001-72	3.6	1.2	8.4	2.9	5.4	49.5	229	24.3	.5	5.8	1.9	62	16.0	96.8	14.1	18.8	37.1	3.93	16.6	2.7	.58	2.11	.35	2.05	.40	1.11	.16	1.18	.15
001-73	4.2	.4	6.6	1.1	2.8	10.4	440	50.9	.3	2.4	2.0	120	15.8	32.9	2.7	7.1	12.4	1.36	5.6	.9	.34	.57	.08	.49	.07	.23	<.05	.32	.05
001-74	3.1	.8	8.5	1.3	3.4	33.9	50	9.5	.2	2.8	1.3	99	13.2	40.6	10.5	14.0	22.1	2.47	10.5	1.9	.45	1.63	.26	1.80	.32	.85	.10	.72	.08
001-75	1.0	.9	8.1	1.5	4.5	37.8	46	7.9	.3	4.0	1.7	129	16.8	48.2	10.8	15.3	26.3	2.83	12.5	2.0	.44	1.68	.27	1.80	.30	.88	.10	.65	.10
001-76	4.3	.7	10.0	1.3	3.3	28.9	159	7.0	.2	2.7	4.2	217	23.8	41.9	5.7	11.9	19.9	2.10	8.6	1.3	.34	.91	.14	.91	.18	.51	.07	.50	.07
RE 001-76	4.5	.9	10.1	1.3	3.4	29.9	148	8.1	.2	3.2	4.4	222	24.1	46.3	5.7	12.0	19.7	2.07	8.3	1.4	.37	.90	.15	.94	.17	.48	.07	.52	.07
001-77	15.5	<.1	4.3	.7	2.4	.8	20	2.6	.1	2.3	1.4	12	2.2	22.4	12.1	9.4	16.7	2.01	8.8	1.8	.47	2.16	.33	1.99	.36	.95	.12	.79	.12
001-78	97.9	.2	4.7	<.5	1.3	7.7	14	305.3	<.1	.3	.2	23	<1	8.1	11.5	6.0	10.7	1.45	6.4	1.8	1.31	1.79	.30	1.96	.35	.78	.09	.54	.08
001-79	1.5	.5	4.0	.7	1.4	9.5	344	15.9	<.1	1.4	3.9	70	21.2	20.1	2.3	4.3	7.5	.80	3.5	.5	.23	.45	.06	.34	.05	.19	<.05	.22	.03
001-80	2.0	.6	7.3	.6	2.3	21.2	557	9.3	.1	1.8	4.4	129	30.1	24.2	3.4	8.0	14.6	1.54	6.2	.7	.39	.61	.08	.55	.12	.37	<.05	.32	.05
001-81	2.8	<.1	4.3	4.3	2.9	5.3	6	24.4	.2	6.9	1.3	19	<1	142.3	8.0	14.5	29.5	2.99	12.8	2.1	.28	1.48	.23	1.24	.25	.73	.10	.74	.12
001-82	8.8	.5	4.6	1.4	3.8	29.3	34	6.8	.3	3.1	1.3	71	6.0	45.8	11.5	18.1	33.2	3.56	15.8	2.6	.39	1.98	.29	1.87	.35	.89	.12	.85	.13
001-83	10.8	.6	4.6	1.6	3.7	27.6	37	7.0	.3	3.2	1.7	64	6.2	45.9	11.9	18.1	33.5	3.68	15.7	2.6	.47	2.08	.31	1.92	.36	.96	.13	.86	.14
001-84	12.3	.3	1.6	<.5	.9	7.8	81	4.2	<.1	1.1	.4	26	2.2	9.9	2.2	3.7	5.9	.66	2.7	.4	.10	.46	.08	.38	.07	.21	<.05	.19	.03
001-85	15.3	6.2	32.5	4.4	18.5	244.7	4	84.3	1.4	24.4	5.5	126	4.4	125.5	40.2	73.1	146.9	15.52	66.2	11.3	2.09	8.43	1.25	7.17	1.33	3.77	.54	3.48	.54
001-86	<.5	.3	2.3	.6	1.5	10.6	<1	1058.4	.1	1.6	1.0	5	<1	21.9	6.0	8.6	17.5	1.75	8.3	1.2	.28	1.17	.16	1.00	.18	.48	.07	.47	.07
001-87	37.6	.8	7.7	1.9	9.4	27.9	21	60.2	.5	6.6	2.8	49	5.7	59.9	19.8	23.5	53.4	5.64	25.5	4.7	1.02	4.47	.56	3.36	.61	1.78	.25	1.43	.23
001-88	7.4	3.2	13.8	2.3	7.0	97.9	<1	49.8	.6	7.6	3.6	383	2.3	77.7	16.2	27.1	51.5	5.69	24.5	3.8	.77	3.18	.44	2.80	.47	1.44	.19	1.40	.19
001-89	<.5	1.0	4.3	<.5	1.6	31.0	1	17.0	.1	1.4	.9	161	2.6	15.0	11.3	8.8	8.9	1.66	7.7	1.1	.24	1.27	.18	1.18	.27	.86	.12	.75	.12
001-90	27.4	.9	22.9	3.4	31.3	28.9	176	372.5	1.8	4.3	4.0	180	8.4	128.2	20.7	18.6	41.1	4.76	22.5	4.3	1.49	4.09	.58	3.56	.69	1.90	.25	1.91	.30
001-91	10.8	5.8	24.5	6.8	14.9	184.8	3	98.9	1.1	18.6	5.8	260	5.0	209.7	33.4	58.3	115.8	12.19	52.9	8.7	1.58	7.05	.90	5.35	1.05	3.07	.43	2.86	.44
001-92	4.9	1.6	8.1	6.6	6.0	42.9	1	104.6	.4	8.4	5.0	178	2.8	201.0	18.8	20.3	39.6	4.23	18.2	3.4	.65	3.22	.47	3.15	.58	1.71	.24	1.67	.25
001-93	54.5	.4	9.3	1.2	5.8	21.8	39	20.8	.4	3.7	1.8	31	4.5	48.6	13.7	18.4	37.0	3.98	18.5	3.2	.32	2.71	.38	2.28	.43	1.12	.16	.93	.15
001-94	26.5	1.7	12.2	2.1	6.6	53.3	2	49.5	.5	7.7	1.7	57	4.0	65.0	12.7	11.9	24.8	2.47	10.9	2.0	.40	1.79	.29	1.95	.43	1.24	.17	1.29	.19
001-95	17.4	3.4	19.3	5.8	14.3	79.7	2	298.9	.9	16.2	10.8	40	4.1	180.9	17.6	30.2	63.7	6.87	29.5	5.0	.91	3.96	.48	3.22	.62	2.34	.33	2.57	.38
STANDARD SO-17	19.2	4.0	20.3	12.8	25.8	22.1	8	314.3	4.2	11.7	13.3	122	11.3	351.2	27.1	10.9	23.8	2.89	14.3	3.3	1.06	3.86	.62	4.19	.95	2.79	.40	2.85	.44

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data ___ FA ___

ACME ANALYTICAL

Barker Minerals Limited FILE # A200561 Page 4 (a)

ACME ANALYTICAL

SAMPLE#	Co ppm	Cs ppm	Ga ppm	Hf ppm	Nb ppm	Rb ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm	W ppm	Zr ppm	Y ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tm ppm	Yb ppm	Lu ppm
001-86	2.5	.3	1.7	.6	1.6	11.7	<1	1084.3	<.1	1.5	1.0	9	<1	22.9	6.1	7.9	15.7	1.84	6.6	1.4	.32	1.12	.17	.88	.18	.50	.07	.44	.05
001-87	43.3	.7	7.3	1.4	8.9	27.9	30	60.7	.6	7.0	3.3	47	5.8	56.8	19.9	23.3	51.8	5.84	22.6	5.1	1.04	4.10	.63	3.31	.63	1.71	.23	1.62	.23
001-88	11.2	3.2	13.1	2.7	8.7	107.9	<1	52.9	.6	8.3	4.3	423	2.5	80.5	16.6	28.1	53.6	6.15	22.4	4.6	.80	3.34	.49	2.80	.55	1.43	.20	1.57	.20
001-89	.6	1.1	4.0	.5	1.4	31.3	<1	17.5	.1	1.2	.9	155	2.5	13.8	11.6	7.8	7.6	1.56	6.2	1.1	.19	.95	.17	1.11	.27	.97	.11	.87	.12
001-90	33.0	.8	21.0	3.5	33.3	31.1	221	395.6	2.0	3.9	5.6	186	8.0	130.1	21.7	19.1	41.8	5.27	20.9	4.6	1.53	4.32	.62	3.49	.71	1.94	.28	1.83	.27
001-91	13.3	5.4	22.1	6.5	14.6	175.2	2	97.0	1.0	18.3	5.5	256	4.8	202.3	31.1	54.8	107.6	11.98	45.3	8.7	1.42	6.52	.89	4.78	.96	2.70	.39	2.77	.38
001-92	8.4	1.6	7.3	6.5	5.8	42.7	<1	104.6	.5	9.3	4.9	178	2.5	202.5	16.6	18.8	38.2	4.28	15.9	3.3	.63	2.98	.45	2.61	.54	1.56	.22	1.59	.21
001-93	55.4	.6	9.1	1.1	6.2	22.6	35	22.2	.4	3.4	1.7	31	4.3	46.3	13.3	18.1	37.6	4.30	17.1	3.5	.33	3.01	.41	2.24	.37	1.19	.13	.87	.13
001-94	39.4	1.6	12.6	2.2	7.3	54.3	1	54.9	.6	8.3	2.2	56	102.6	67.4	13.6	12.4	22.9	2.63	9.9	2.2	.44	1.90	.33	2.21	.44	1.31	.16	1.32	.18
001-95	20.7	3.3	18.4	5.5	15.2	84.1	<1	319.0	1.0	15.7	13.6	39	2.7	172.8	18.5	30.6	64.4	7.51	27.5	6.2	.99	4.16	.52	3.24	.67	2.23	.36	2.65	.38
001-96	14.1	3.3	13.9	6.1	9.1	44.1	<1	297.2	.7	15.0	4.5	88	10.1	200.0	24.0	40.8	80.9	8.98	33.7	7.5	1.15	5.48	.73	3.75	.79	2.26	.33	2.36	.31
RE 001-96	14.3	3.5	13.7	6.8	9.6	47.3	<1	303.0	.8	16.3	4.1	96	11.1	215.8	24.2	41.4	82.6	9.13	35.8	7.2	1.21	5.44	.73	4.00	.78	2.26	.33	2.43	.33
001-97	56.9	6.1	20.0	4.4	12.9	177.1	2	49.4	1.0	13.6	6.7	151	2.8	124.4	27.7	18.7	42.1	4.85	18.6	4.1	.77	3.52	.65	4.24	.92	2.87	.42	3.10	.45
001-98	7.2	4.4	20.4	5.4	14.6	180.3	6	62.7	1.3	18.3	5.9	126	2.0	157.9	27.9	44.4	91.6	10.21	37.9	8.1	1.26	6.20	.83	4.54	.85	2.51	.33	2.35	.34
001-99	14.4	2.0	13.7	5.6	9.8	110.9	12	39.9	.7	13.3	5.7	83	3.3	173.9	22.0	36.4	73.6	8.15	30.0	6.1	1.03	4.40	.61	3.59	.71	2.17	.28	2.06	.28
001-100	56.0	1.2	11.0	2.7	18.1	50.8	3	267.3	1.1	2.2	1.2	157	2.1	94.6	15.2	15.8	31.9	3.85	15.5	4.0	.94	3.37	.48	2.67	.55	1.35	.17	1.20	.17
001-101	15.4	2.2	15.7	4.9	10.6	132.6	3	51.7	.8	13.1	5.2	134	4.6	159.5	31.7	35.3	70.3	7.70	30.5	6.0	1.08	5.09	.74	4.63	.92	2.73	.40	2.64	.39
001-102	5.0	1.6	6.3	.8	2.4	43.9	4	44.1	.2	2.8	4.1	258	1.9	26.5	23.9	13.7	15.3	2.89	12.1	2.9	.61	2.89	.43	2.45	.60	1.73	.21	1.42	.23
001-103	7.3	<.1	<.5	<.5	<.5	1.0	653	22.3	<.1	.1	.4	5	1.2	1.9	.4	.8	1.0	.13	.4	.1	.08	.06	<.01	.08	<.05	<.05	<.05	<.05	<.01
001-104	6.8	3.0	10.8	2.2	7.5	92.6	4	43.7	.5	7.1	4.4	379	1.6	77.8	17.6	23.8	43.9	5.12	19.9	3.8	.82	2.96	.44	2.51	.49	1.50	.21	1.45	.22
001-105	98.0	<.1	1.7	<.5	.6	.8	9	3.5	<.1	<.1	.2	34	<1	1.6	1.4	1.6	2.5	.32	1.4	.3	<.05	.19	.03	.23	<.05	.11	<.05	.14	.02
001-106	26.1	.8	15.8	2.4	3.3	36.6	<1	257.5	.2	1.8	1.1	260	<1	71.0	20.0	11.7	27.3	3.20	14.2	4.0	1.11	3.52	.57	3.37	.65	1.99	.27	1.86	.33
001-107	34.0	.3	<.5	<.5	<.5	1.4	<1	5.8	<.1	.4	.1	<5	3.5	4.9	.6	1.9	2.4	.19	.9	.2	<.05	.16	.02	.13	<.05	.06	<.05	.09	<.01
001-108	90.1	.1	2.0	<.5	1.1	<.5	<1	6.2	.2	.1	.9	36	<1	.7	6.0	1.4	2.2	.33	1.5	.6	.07	.87	.13	.78	.18	.48	.07	.52	.07
001-109	88.6	<.1	1.2	<.5	<.5	<.5	<1	.9	<.1	<.1	.3	34	<1	.6	.8	.8	1.1	.09	<.4	<.1	<.05	.16	<.01	.13	<.05	.08	<.05	.07	.01
001-110	12.7	.2	2.1	1.3	2.1	13.4	<1	13.4	.1	.7	.2	9	2.6	40.3	4.6	2.8	5.5	.52	2.2	.4	.14	.55	.08	.66	.15	.42	.06	.39	.06
001-111	56.5	.9	17.8	2.8	15.6	38.6	103	51.8	1.0	4.7	7.9	236	28.0	92.4	48.2	25.4	51.0	5.97	25.3	6.3	1.21	6.14	1.07	6.43	1.41	3.81	.52	3.42	.47
001-112	75.1	.5	16.7	2.5	15.6	25.2	134	37.9	1.0	1.6	3.5	247	44.2	77.4	44.2	12.1	26.1	3.10	14.2	4.5	.90	6.14	1.03	6.61	1.35	3.61	.48	3.26	.48
001-113	47.5	1.4	16.1	3.3	15.5	54.3	61	51.3	.9	6.0	11.6	232	14.2	111.6	34.5	22.6	47.7	5.54	22.9	5.3	1.24	5.43	.84	4.97	1.00	2.75	.36	2.54	.40
001-114	9.9	.7	9.0	.9	3.9	33.8	35	34.2	.3	3.5	5.0	121	4.6	44.6	20.4	17.8	35.5	3.81	14.0	2.9	.61	2.61	.43	2.62	.61	1.66	.23	1.61	.24
001-115	141.6	.6	7.4	1.7	8.7	32.1	32	51.7	.6	6.6	2.7	46	5.1	51.9	27.0	23.8	54.3	5.66	22.6	5.4	1.13	4.98	.80	4.26	.86	2.35	.31	2.13	.30
001-116	21.4	.7	5.8	2.0	4.5	37.4	27	12.1	.3	4.6	5.7	109	4.4	63.4	8.3	19.1	33.4	3.56	13.4	2.6	.45	2.06	.22	1.18	.24	.74	.10	.87	.11
STANDARD SO-17	18.9	3.8	20.0	12.1	25.9	23.9	12	319.0	4.3	11.6	12.8	124	11.2	351.2	27.4	11.0	23.6	3.01	13.5	3.3	1.09	3.93	.66	4.15	.93	2.92	.41	2.94	.43

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

SAMPLE#	Co ppm	Cs ppm	Ga ppm	Hf ppm	Nb ppm	Rb ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm	W ppm	Zr ppm	Y ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tm ppm	Yb ppm	Lu ppm
001-117	15.0	3.5	24.0	6.5	15.1	174.8	14	62.2	1.1	17.1	4.7	75	4.2	235.0	25.5	52.6	107.1	12.53	47.3	7.5	1.32	6.25	.86	4.68	.85	2.48	.35	2.40	.38
001-118	16.8	.2	1.4	<.5	1.3	8.8	14	3.5	<.1	1.1	2.3	13	4.2	15.1	2.3	3.5	6.6	.74	2.8	.5	.11	.48	.08	.42	.08	.21	<.05	.29	.03
001-119	4.0	<.1	3.0	<.5	2.2	6.2	<1	101.6	<.1	.1	.2	30	1.2	9.7	6.0	2.0	4.2	.65	3.0	.8	.49	1.24	.16	1.11	.21	.50	.07	.37	.05
001-120	48.9	1.8	21.2	3.7	25.4	60.3	2	204.8	1.6	3.0	1.1	265	3.8	150.3	25.8	19.7	41.9	5.44	24.7	5.2	1.52	5.83	.84	4.72	.95	2.44	.34	1.86	.26
001-121	18.5	1.0	13.0	5.1	13.7	33.8	1	76.9	1.0	5.9	1.0	122	3.5	193.5	15.4	18.3	37.0	4.47	17.6	3.1	.93	2.95	.48	2.73	.54	1.57	.23	1.43	.20
001-122	60.3	1.7	7.8	1.1	6.7	5.3	2	338.6	.5	.9	.4	111	1.7	47.2	8.8	6.0	13.7	1.76	8.4	1.8	.55	1.91	.29	1.68	.34	.95	.10	.66	.10
001-123	46.5	.4	20.8	3.6	29.4	9.7	2	639.4	2.0	3.0	.8	258	<1	141.6	23.4	25.3	49.8	6.35	26.3	4.6	1.66	5.46	.72	4.24	.84	2.27	.30	1.92	.23
001-124	94.1	.2	1.0	<.5	<.5	.6	<1	4.8	<.1	<.1	<.1	29	<1	1.0	.8	<.5	1.3	.19	.7	.1	<.05	.14	.03	.16	<.05	.07	<.05	.06	<.01
001-125	23.5	.4	2.8	1.1	2.5	19.3	3	8.2	.1	2.6	2.1	26	5.0	37.8	4.5	6.8	13.0	1.59	6.2	1.0	.17	.94	.16	.75	.16	.48	.06	.36	.06
001-126	34.9	1.1	17.6	1.4	2.4	38.5	<1	547.9	.2	1.5	1.0	247	<1	51.5	18.0	8.4	18.4	2.60	12.5	2.9	.98	3.34	.53	3.19	.66	1.79	.27	1.71	.26
RE 001-126	33.4	1.2	17.0	1.4	2.4	37.5	<1	534.2	.2	1.6	1.0	239	<1	51.0	17.8	8.5	16.8	2.51	12.7	2.7	.91	3.08	.52	3.21	.63	1.87	.27	1.67	.23
STANDARD SO-17	18.2	3.5	20.1	12.3	25.4	22.7	8	309.3	4.3	11.3	11.8	120	8.9	365.8	26.5	10.5	23.2	3.03	13.5	3.0	1.01	3.98	.68	4.39	.98	2.88	.41	2.80	.43

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

ACME ANALYTICAL LABORATORIES LTD. 852 E. HASTINGS ST. VANCOUVER BC V6A 1R6 PHONE(604)253-3158 FAX(604)253-1716
(ISO 9002 Accredited Co.)

GEOCHEMICAL ANALYSIS CERTIFICATE

Barker Minerals Limited File # A200561 Page 1 (b)

22117 - 37A Avenue, Langley BC V2Z 1N9 Submitted by: Louis Doyle

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ni ppm	As ppm	Cd ppm	Sb ppm	Bi ppm	Ag ppm	Au ppb	Hg ppm	Tl ppm	Se ppm	Te ppm	Au** ppb	Pt** ppb	Pd** ppb
001-01	.5	126.5	1.0	55	5.9	1.1	.1	.1	.1	.1	26.1	.11	.1	<1	<1	36	<2	<2
001-02	.8	38.1	1.3	51	47.3	1.2	<.1	.1	.1	<.1	7.6	.01	<.1	<1	<1	18	<2	<2
001-03	1.8	297.7	4.9	62	11.8	10.1	.2	.1	.1	.1	20.0	.10	<.1	<1	<1	34	2	11
001-04	.7	94.7	4.4	64	36.0	13.5	.1	.2	<.1	<.1	5.7	.02	<.1	<1	<1	4	5	6
001-05	.4	76.1	3.4	76	23.1	7.9	.1	.5	<.1	.1	2.6	.06	<.1	<1	<1	3	6	7
001-06	.6	76.9	3.3	62	15.7	8.2	.1	.3	<.1	.1	3.1	.08	<.1	<1	<1	5	4	7
001-07	1.1	33.3	.7	30	8.8	.7	.1	.1	<.1	<.1	<.5	.04	<.1	<1	<1	<2	3	2
001-08	1.5	57.5	6.5	83	11.1	61.7	.1	1.0	.1	.1	8.2	.04	<.1	2	<1	7	<2	<2
001-09	1.3	4009.8	283.9	386	91.4	4484.4	2.1	6.4	23.0	14.5	32.6	.04	<.1	4	<1	57	<2	<2
001-10	3.4	2417.2	241.8	2660	110.1	363.4	10.1	.9	37.1	5.0	25.9	.14	.1	26	<1	47	<2	<2
RE 001-10	3.3	2361.4	240.9	2616	107.7	377.5	9.5	.9	37.5	4.9	34.3	.14	.1	25	<1	48	<2	<2
001-11	.9	1097.6	160.8	190	117.5	3.7	<.1	.2	13.0	1.7	22.9	.01	<.1	4	1	35	<2	<2
001-12	1.5	793.6	17853.8	88955	241.0	6.6	266.5	11.9	60.2	33.3	35.0	3.38	.1	34	1	60	<2	4
001-13	.9	680.5	415.1	2207	147.2	<.5	8.9	.3	8.7	1.7	6.8	.08	<.1	4	1	12	<2	2
001-14	1.2	127.8	19858.9	31829	40.3	1.1	123.7	1.4	237.6	68.7	22.0	1.58	.5	25	4	63	<2	<2
001-15	1.3	22.5	127.7	180	44.8	12.6	.5	.1	1.6	.4	1.0	.02	.1	<1	<1	<2	<2	<2
001-16	2.3	72.1	20.2	33	3.5	3.5	.1	.1	.7	.2	3.8	.02	<.1	<1	<1	10	<2	<2
001-17	1.8	363.8	248.0	319	19.0	47.8	1.0	.3	1.9	1.1	76.4	.05	<.1	2	<1	106	<2	<2
001-18	7.5	824.1	218.7	699	136.9	2738.5	2.9	2.4	8.2	2.0	5.0	.05	.1	9	<1	14	2	6
001-19	6.9	971.1	328.2	1789	127.8	815.4	3.8	1.4	5.0	2.1	2.9	.13	.1	3	<1	5	2	2
001-20	11.8	1159.6	122.7	61	125.4	2.8	.2	.5	2.5	1.1	3.3	.01	.1	2	<1	5	3	2
001-21	2.3	586.4	143.5	235	180.7	32.3	.4	.2	3.7	1.1	2.2	.03	<.1	4	<1	4	<2	<2
001-22	3.6	120.7	10.8	15	11.6	31.1	<.1	.1	1.3	.2	1.6	<.01	<.1	1	<1	3	<2	<2
001-23	1.5	52.9	14.5	22	4.0	25.1	<.1	.1	4.3	.1	3.4	<.01	<.1	<1	<1	5	<2	<2
001-24	.8	728.7	18.1	89	27.4	<.5	<.1	.1	2.9	.3	2.7	.01	<.1	<1	<1	3	<2	<2
001-25	1.7	952.0	12113.2	37682	224.9	6.3	142.6	4.0	24.4	12.5	59.2	1.74	.1	20	<1	80	<2	3
001-26	9.5	2615.0	454.9	539	192.1	6.6	1.7	.6	10.7	4.3	8.8	.02	.1	9	<1	5	2	3
001-27	1.1	1559.0	264.7	132	288.4	3.6	.1	.1	20.0	2.6	9.9	<.01	<.1	9	2	19	<2	<2
001-28	1.9	141.5	153.6	248	22.1	37.5	.6	.2	5.3	.4	7.4	.03	<.1	<1	<1	6	<2	<2
001-29	3.2	95.9	9.1	14	9.5	43.4	<.1	.5	2.2	.2	14.1	.03	<.1	1	<1	36	<2	<2
001-30	7.3	502.0	176.9	120	27.9	133.5	.2	2.8	14.7	1.2	12.4	.04	.1	30	<1	44	<2	3
STANDARD DS3/FA-10R	9.1	124.7	34.1	151	35.9	29.1	5.3	4.9	5.4	.3	23.8	.21	1.1	1	1	502	470	498

GROUP 1DX - 0.50 GM SAMPLE LEACHED WITH 3 ML 2-2-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR, DILUTED TO 10 ML, ANALYSED BY ICP-MS.
UPPER LIMITS - AG, AU, HG, W = 100 PPM; MO, CO, CD, SB, BI, TH, U & B = 2,000 PPM; CU, PB, ZN, NI, MN, AS, V, LA, CR = 10,000 PPM.
- SAMPLE TYPE: ROCK R150 60C AU** PT** & PD** GROUP 3B BY FIRE ASSAY & ANALYSIS BY ICP-ES. (30 gm)
Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

DATE RECEIVED: MAR 5 2002 **DATE REPORT MAILED:** March 22/02 **SIGNED BY** ....D. TOYE, C.LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

Assay recommend for Cu, Pb, Zn >1%

 Data FA ____

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ni ppm	As ppm	Cd ppm	Sb ppm	Bi ppm	Ag ppm	Au ppb	Hg ppm	Tl ppm	Se ppm	Te ppm	Au** ppb	Pt** ppb	Pd** ppb
001-31	1.3	1990.1	67.6	59	130.4	.7	.2	.2	20.9	.7	6.9	<.01	<.1	3	<1	8	<2	<2
001-32	.6	211.4	10.7	87	37.5	<.5	<.1	.1	2.6	.1	2.1	.01	<.1	<1	<1	2	<2	<2
001-33	8.5	730.3	210.4	763	165.5	1483.5	1.6	1.4	5.3	1.8	6.5	.04	.1	4	<1	10	2	5
001-34	5.4	327.7	1674.2	414	316.8	353.6	2.8	1.2	29.0	11.3	6.0	.02	.1	12	1	13	4	9
001-35	2.3	21947.0	1550.7	323	30.8	4527.9	2.3	55.5	47.4	74.9	131.5	.74	.4	112	<1	854	<2	<2
001-36	2.4	2170.8	18.1	81	58.7	11.7	.6	.4	1.1	1.4	26.2	.01	<.1	<1	<1	16	4	<2
001-37	9.9	1136.8	472.5	4736	174.4	382.9	11.8	1.2	7.8	3.1	5.0	.30	<.1	4	<1	16	<2	3
001-38	4.6	1258.0	83.0	32	131.3	216.4	.2	2.9	5.4	1.3	122.4	<.01	<.1	118	1	159	<2	11
001-39	14.2	1874.3	7034.2	6499	34.5	10632.9	20.6	151.3	73.8	83.7	124.9	44.00	14.7	43	<1	129	<2	<2
001-39A	12.1	1248.6	7998.4	4382	29.4	5021.3	14.2	73.6	27.3	34.0	58.9	26.00	8.4	15	<1	61	<2	<2
001-40	8.6	3007.8	1085.8	2442	144.0	1009.9	5.9	1.4	17.7	7.7	7.4	.22	.2	4	<1	7	<2	2
RE 001-40	8.0	2870.9	1039.9	2360	138.4	956.0	5.3	1.3	17.1	7.1	7.1	.24	.1	5	<1	7	4	5
001-41	3.9	786.5	4838.5	2988	19.2	868.7	9.5	27.2	24.2	17.3	26.8	18.00	3.3	7	<1	46	<2	<2
001-42	11.2	1314.8	7771.5	6710	36.2	1580.7	20.1	61.7	15.3	36.6	34.5	36.00	5.5	12	<1	37	<2	<2
001-43	21.8	2762.4	1531.9	146	37.2	2184.5	.6	103.0	91.9	52.8	24.8	13.00	26.3	38	<1	144	<2	<2
001-44	4.4	948.4	89.5	20	269.4	13.3	.1	.4	130.6	1.1	5.4	.03	.2	23	<1	6	<2	<2
001-44A	5.9	2270.1	65.4	45	194.5	32003.9	.2	15.7	63.9	1.3	52.2	.03	.1	22	<1	76	4	59
001-44B	3.3	12309.5	130.5	199	177.8	767.6	1.0	1.5	172.0	4.6	22.7	.09	.1	20	<1	37	<2	4
001-45	1.8	83.1	11.4	91	28.1	50.9	.3	.1	1.5	.1	1.9	.01	.1	<1	<1	2	<2	<2
001-46	1.0	20746.5	1930.3	1280	47.6	716.1	2.8	4.1	347.2	129.0	82.0	.45	<.1	34	1	150	<2	<2
001-47	2.1	95.3	43.7	41	22.4	20.3	.1	.2	1.9	.6	4.1	.02	.1	<1	<1	<2	<2	<2
001-48	1.1	56.9	22.2	58	24.4	8.4	.1	.2	1.2	.3	1.0	.04	.1	<1	<1	2	<2	<2
001-55	.6	70.7	8.4	32	43.9	5.6	.1	.2	.2	.1	.6	.01	<.1	<1	<1	2	<2	2
001-56	.8	118.9	4.3	56	31.8	2.9	<.1	.1	.1	<.1	1.3	.02	.2	<1	<1	2	7	9
001-57	.4	74.3	6.4	42	28.2	4.0	.1	.2	.5	.2	4.4	.01	<.1	<1	<1	11	4	3
001-58	3.3	261.5	6.0	50	32.6	5.0	.1	.2	.4	.2	1.7	.01	<.1	<1	<1	6	4	4
001-59	.6	48.2	1.3	38	25.4	1.1	.1	.1	<.1	<.1	1.2	.01	<.1	<1	<1	2	2	2
001-60	1.7	83.5	4.3	53	42.0	2.6	<.1	.2	.3	.1	1.6	.01	<.1	<1	<1	4	3	3
001-61	.6	16.5	2.7	33	45.3	1.3	.1	.1	<.1	<.1	1.3	<.01	<.1	<1	<1	<2	<2	2
001-62	4.1	2948.3	1708.7	281	28.7	1528.8	1.1	58.6	35.0	22.5	2.7	2.31	3.9	21	<1	251	<2	<2
001-63	2.2	44780.9	2442.7	6553	24.4	641.1	24.3	210.7	680.5	104.4	43.9	6.14	2.5	126	1	670	2	2
001-64	1.1	478.7	29.0	84	793.5	508.3	.3	2.1	5.4	.9	3.0	.04	.1	<1	<1	7	5	6
STANDARD DS3/FA-10R	9.0	119.3	32.3	143	32.6	28.0	5.2	4.7	5.2	.3	21.1	.26	1.0	1	1	480	459	464

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ni ppm	As ppm	Cd ppm	Sb ppm	Bi ppm	Ag ppm	Au ppb	Hg ppm	Tl ppm	Se ppm	Te ppm	Au** ppb	Pt** ppb	Pd** ppb
001-65	3.9	4403.0	1770.0	774	17.0	1368.1	3.4	36.9	68.5	46.3	35.1	.97	2.1	46	<1	308	<2	<2
001-66	1.9	9661.0	261.1	406	103.9	49.4	1.8	1.1	2.6	13.8	83.4	.07	.8	2	<1	116	<2	<2
001-67	5.6	184.3	5307.9	72848	12.8	166.3	246.9	66.8	96.0	65.7	55.1	909.00	1.1	30	<1	61	<2	<2
001-68	19.0	3268.0	3241.7	1675	48.4	1193.4	8.5	1510.3	61.6	95.8	8.4	51.00	7.9	34	<1	211	<2	<2
001-69	.8	622.6	113.9	2266	612.4	92.8	13.6	6.7	2.1	1.2	8.8	.36	.1	<1	<1	9	3	4
001-70	8.9	5319.3	1630.1	800	35.8	1226.9	3.4	297.2	51.3	89.3	1.0	24.00	34.8	23	<1	132	<2	<2
001-71	6.8	562.1	5818.7	6748	23.7	2630.7	26.1	62.6	9.8	27.2	38.3	41.00	9.9	15	<1	39	<2	<2
001-72	4.6	584.2	7956.9	5117	21.5	3925.9	17.7	72.8	33.0	36.0	79.0	67.00	2.3	10	<1	84	<2	<2
001-73	6.7	547.5	11309.4	7737	15.0	8312.4	28.3	151.9	16.3	49.9	111.5	147.00	4.1	21	<1	111	2	2
001-74	6.0	4491.1	4229.6	7379	14.4	29528.7	23.7	102.9	108.6	44.9	123.8	55.00	1.4	177	1	156	<2	<2
001-75	6.1	5328.6	3779.1	4521	22.3	12115.5	15.3	70.3	63.1	26.8	114.9	37.00	3.4	126	<1	139	<2	<2
001-76	9.8	4614.2	2693.4	3949	34.7	15214.0	16.6	138.8	85.5	39.2	24.5	41.00	7.9	160	<1	172	<2	<2
RE 001-76	13.0	4729.7	2703.0	3983	35.7	15489.3	16.1	139.8	84.9	39.2	70.0	40.00	8.0	166	<1	177	<2	<2
001-77	1.5	518.7	188.0	157	146.7	231.9	.4	1.7	10.7	3.0	5.8	.82	.1	19	<1	5	<2	<2
001-78	.7	339.9	157.4	65356	90.0	324.3	640.8	2.4	1.2	1.1	21.3	3.93	.1	1	<1	29	<2	<2
001-79	31.4	3976.6	2984.8	1321	68.5	1605.8	7.6	133.1	78.0	40.5	5.5	35.00	42.9	38	<1	185	<2	<2
001-80	18.3	4330.9	3107.7	4164	34.4	28314.2	17.5	331.2	68.8	50.1	72.1	52.00	20.0	59	<1	185	<2	<2
001-81	1.3	1148.2	125.8	344	54.7	1306.8	2.6	2.2	3.6	2.7	8.9	.42	.2	2	<1	7	2	2
001-82	2.4	6922.2	1850.7	492	28.4	727.8	2.2	17.9	117.8	36.0	2.2	.97	.4	111	<1	69	<2	<2
001-83	3.6	5838.1	1806.6	487	29.2	884.7	2.1	20.7	110.1	34.7	1.5	1.17	1.0	102	<1	84	<2	<2
001-84	5.4	13139.8	1551.8	630	29.2	2568.3	3.5	79.5	76.1	41.5	7.0	1.58	6.0	32	<1	179	<2	<2
001-85	1.0	99.9	23.6	81	42.0	15.7	.3	.3	.7	.3	<.5	.09	.1	<1	<1	<2	<2	<2
001-86	.4	27.3	49.4	10	1.8	11.7	.1	.4	.7	.2	<.5	.05	<.1	<1	<1	13	<2	<2
001-87	7.3	2799.0	9882.4	27304	132.9	370.5	93.7	1.3	24.3	12.8	13.1	2.00	<.1	27	<1	45	<2	<2
001-88	9.6	60.9	90.6	398	42.2	15.7	3.2	.4	.5	.4	1.3	.09	.1	3	<1	3	<2	<2
001-89	4.0	14.0	55.8	44	6.2	9.9	.3	.6	.2	.5	2.2	.18	<.1	2	<1	16	2	3
001-90	5.4	457.4	14.3	86	62.2	.8	.1	.3	11.8	.1	49.8	.02	<.1	6	<1	56	<2	<2
001-91	8.2	33.7	26.9	200	56.2	22.4	1.3	.4	.5	.3	3.0	.04	.1	2	<1	<2	<2	<2
001-92	5.5	57.2	13.7	198	31.2	9.3	1.7	.2	.1	.3	.5	.05	.1	5	<1	11	<2	<2
001-93	.9	1318.2	1445.3	706	31.3	3218.9	1.1	7.8	24.7	10.8	29.2	.22	<.1	16	<1	37	<2	<2
001-94	2.6	151.9	167.6	16	48.8	17.5	.3	.2	186.5	1.0	1907.4	.02	<.1	<1	2	2731	<2	<2
001-95	39.1	100.4	48.3	49	31.6	7.9	<.1	.1	82.6	.5	820.1	.02	.1	<1	2	803	3	2
STANDARD DS3/FA-10R	9.7	120.5	33.5	146	35.7	28.8	5.2	4.9	5.6	.3	21.0	.24	1.1	<1	1	476	461	462

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA ____

ACME ANALYTICAL

ACME ANALYTICAL

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ni ppm	As ppm	Cd ppm	Sb ppm	Bi ppm	Ag ppm	Au ppb	Hg ppm	Tl ppm	Se ppm	Te ppm	Au** ppb	Pt** ppb	Pd** ppb
001-86	.4	21.6	44.8	10	3.3	10.8	.1	.4	.6	.2	<.5	<.01	<.1	1	<1	11	<2	<2
001-87	8.9	4182.0	14585.0	44590	194.0	480.3	109.7	1.4	27.2	14.9	20.5	1.33	<.1	41	<1	109	<2	<2
001-88	11.2	68.5	107.6	419	46.4	17.6	3.6	.5	.6	.4	<.5	.04	.1	4	<1	3	<2	<2
001-89	4.1	12.4	53.3	41	5.9	9.0	.2	.7	.2	.5	.9	.15	<.1	1	<1	21	<2	3
001-90	5.4	408.8	14.5	78	58.7	.7	.1	.2	11.6	.1	46.6	.01	<.1	6	<1	60	<2	<2
001-91	8.6	30.5	24.5	187	64.2	20.4	1.1	.4	.5	.2	1.8	.04	.1	3	<1	7	<2	<2
001-92	5.7	56.0	13.6	190	32.2	9.1	1.9	.2	.1	.3	<.5	.02	<.1	7	<1	15	<2	<2
001-93	.9	1231.3	1363.0	606	28.1	2805.0	1.0	6.8	21.8	10.5	25.6	.11	<.1	17	<1	40	<2	<2
001-94	2.9	149.5	172.4	17	50.0	19.0	.3	.2	193.5	1.6	1979.6	.02	<.1	1	2	2997	<2	<2
001-95	38.4	95.5	46.3	45	30.8	6.2	<.1	.1	79.6	.5	882.5	.01	.2	<1	2	876	<2	<2
001-96	2.0	78.6	15.2	53	40.8	2.6	.1	.1	.7	.1	27.6	.02	.2	<1	<1	8	<2	<2
RE 001-96	1.5	79.5	15.1	51	38.5	2.5	.1	.1	.6	.1	14.5	.01	.2	<1	<1	8	<2	<2
001-97	28.2	149.4	33.2	121	117.1	1.0	.2	.1	2.3	.2	6.3	.02	.1	<1	<1	10	2	2
001-98	2.7	50.9	37.5	128	27.6	3.7	.6	.2	.2	.2	5.1	.03	.1	3	<1	4	<2	<2
001-99	3.4	111.1	911.2	1635	44.4	124.2	6.3	2.8	1.3	4.2	21.2	1.80	.4	5	<1	80	<2	<2
001-100	1.2	69.9	8.2	60	348.8	98.0	.2	.3	.1	<.1	3.6	<.01	<.1	<1	<1	7	6	6
001-101	7.5	143.3	117.6	184	59.7	140.4	.5	3.8	.7	.6	13.6	.11	.4	5	<1	31	<2	<2
001-102	8.5	59.7	46.7	156	58.7	13.7	.4	.3	.2	.2	2.4	.02	<.1	1	<1	22	3	4
001-103	2.1	2792.0	1561.0	97	23.8	292.2	.3	48.9	28.2	7.5	2.3	1.51	.3	25	<1	177	<2	<2
001-104	19.5	17.9	48.5	72	40.0	24.3	.1	1.8	.5	.4	1.6	.09	.1	4	<1	<2	<2	<2
001-105	.2	12.6	5.6	12	1537.3	23.7	.1	1.1	.1	<.1	.8	<.01	<.1	<1	<1	4	4	2
001-106	1.0	132.9	3.4	57	18.1	3.3	<.1	.2	.1	.1	1.2	<.01	<.1	1	<1	2	4	4
001-107	2.6	438.5	52.3	9	22.7	2.5	.1	.4	330.3	.6	4.1	.01	<.1	1	11	3	<2	<2
001-108	.1	5.1	2.2	7	1621.3	16.0	<.1	.9	1.6	<.1	1.5	<.01	<.1	<1	<1	<2	6	6
001-109	.1	2.8	2.0	10	1157.4	4.8	<.1	.3	.2	<.1	<.5	<.01	<.1	<1	<1	<2	7	8
001-110	2.1	24.5	1796.4	49	36.9	3027.5	.1	2.3	1.6	2.5	698.6	<.01	<.1	2	<1	743	<2	<2
001-111	3.4	4206.0	321.0	518	94.8	127.9	1.3	.3	9.7	4.9	31.3	<.01	<.1	2	<1	5	3	5
001-112	1.7	8836.8	538.9	951	130.3	214.9	3.0	.5	17.5	11.1	17.9	.01	<.1	5	<1	38	2	2
001-113	4.3	6344.8	115.8	358	124.9	89.2	.7	.2	3.2	1.4	5.6	.01	<.1	<1	<1	4	3	5
001-114	8.6	247.7	309.8	149	24.4	144.9	<.1	.9	11.1	5.5	2.3	.03	<.1	4	<1	32	2	6
001-115	4.7	4936.4	5176.3	24208	128.9	512.5	76.5	.9	15.3	9.2	8.1	1.12	<.1	33	<1	72	<2	<2
001-116	10.4	4530.0	3038.0	1463	91.9	368.3	4.4	3.1	61.7	28.8	5.1	.19	.2	88	<1	92	<2	<2
STANDARD DS3/FA-10R	9.4	119.6	34.8	148	34.6	28.4	5.5	4.8	5.5	.3	22.4	.22	1.2	<1	1	488	481	474

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data ✓ FA ____

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ni ppm	As ppm	Cd ppm	Sb ppm	Bi ppm	Ag ppm	Au ppb	Hg ppm	Tl ppm	Se ppm	Te ppm	Au** ppb	Pt** ppb	Pd** ppb
001-117	1.7	87.9	92.6	123	39.2	144.2	.3	.8	.7	.5	1.3	.04	.3	<1	<1	2	<2	<2
001-118	7.1	5250.0	2779.0	828	21.8	1188.0	3.6	33.0	76.7	60.6	22.3	1.18	2.1	63	<1	325	<2	<2
001-119	2.3	15.3	225.2	29	22.1	15.0	.3	.4	1.1	.7	.7	.03	<.1	1	<1	<2	<2	<2
001-120	1.6	139.2	7.7	117	233.2	57.3	.3	.2	.1	.2	1.1	.02	<.1	<1	<1	4	5	5
001-121	1.5	29.1	6.9	78	69.3	26.4	.1	.1	<.1	.1	<.5	.01	.1	<1	<1	<2	<2	<2
001-122	1.2	61.3	10.6	84	549.4	317.8	.3	.6	.1	.1	1.4	<.01	<.1	<1	<1	3	7	9
001-123	.4	22.8	1.9	31	87.9	1.6	.1	.1	<.1	<.1	.8	<.01	<.1	2	<1	2	2	4
001-124	.1	4.1	1.5	9	1422.2	9.5	.1	.5	.1	<.1	1.1	<.01	<.1	1	<1	3	7	6
001-125	6.2	4817.0	2742.0	944	28.0	1239.0	4.0	22.8	73.9	53.2	21.8	2.37	5.1	55	<1	360	<2	<2
001-126	1.7	118.3	7.2	96	52.6	5.8	.6	.2	.1	.3	1.3	.02	.1	4	<1	<2	6	6
RE 001-126	1.7	119.9	5.9	99	54.5	5.1	.6	.3	.1	.3	.8	.03	<.1	3	<1	4	6	7
STANDARD DS3/FA-10R	9.1	120.3	33.6	151	35.7	29.4	5.7	4.8	5.4	.3	19.7	.22	1.2	1	1	476	465	474

Sample type: ROCK R150 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

Data FA

Eco Tech LABORATORY LTD.

ASSAYING
GEOCHEMISTRY
ANALYTICAL CHEMISTRY
ENVIRONMENTAL TESTING

10041 Dallas Drive, Kamloops, B.C. V2C 6T4
Phone (250) 573-5700 Fax (250) 573-4557
email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-124B

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

2-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 5
Sample type: Core
Project #: Frank Creek
Shipment #: None Given
Samples submitted by: Chris Wild

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
61	08235	<5	<5	<5
62	08236	<5	<5	<5
63	08237	<5	<5	<5
64	08238	<5	10	<5
65	08239	<5	<5	<5

QC DATA:

	Au (ppb)	Pd (ppb)	Pt (ppb)
Standard:			
GEO'02	110	<5	<5

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

JJ/kk
XLS/02
E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

Eco Tech LABORATORY LTD.

ASSAYING
GEOCHEMISTRY
ANALYTICAL CHEMISTRY
ENVIRONMENTAL TESTING

10041 Dallas Drive, Kamloops, B.C. V2C 6T4
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email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-124

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

2-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 60
Sample type: Core
Project #: ACE
Shipment #: None Given
Samples submitted by: Chris Wild

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
1	08151 Ace 02-01 13.0-14.5m	5	<5	5
2	08152 Ace 02-01 17.37-19.0m	<5	<5	5
3	08153 Ace 02-01 29.0-30.5m	<5	<5	5
4	08154 Ace 02-01 37.5-39.0m	5	<5	5
5	08155 Ace 02-01 45.0-46.5m	<5	<5	<5
6	08156 Ace 02-01 47.5-49.0m	5	<5	5
7	08157 Ace 02-01 49.0-49.7m	<5	<5	5
8	08158	5	5	5
9	08159 Ace 02-01 50.7-52.1m	5	<5	5
10	08160 Ace 02-01 52.1-54.0m	<5	<5	<5
11	08161 Ace 02-01 54.0-55.2m	<5	<5	5
12	08162 Ace 02-01 55.2-57.0m	<5	<5	5
13	08163 Ace 02-01 57.0-58.8m	<5	<5	<5
14	08164 Ace 02-01 58.8-61.0m	<5	<5	5
15	08165 Ace 02-01 61.0-63.0m	<5	<5	5
16	08166 Ace 02-01 63.0-65.3m	<5	<5	5
17	08167 Ace 02-01 65.3-67.5m	<5	<5	5
18	08168 Ace 02-01 76.0-77.5m	<5	<5	5
19	08169 Ace 02-01 88.0-89.5m	<5	<5	<5
20	08170 Ace 02-01 103.6-105.1m	<5	<5	5
21	08171 Ace 02-01 130.0-131.5m	<5	<5	5
22	08172 Ace 02-01 139.5-141.0m	<5	<5	<5
23	08173	<5	<5	<5
24	08174	30	<5	5
25	08175	<5	<5	<5
26	08176	<5	<5	5
27	08177	<5	<5	5
28	08178	20	<5	5
29	08179	<5	<5	5

BARKER MINERALS LTD. AK 2002-124 2-Jul-02

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
30	08180	40	10	5
31	08181	25	15	5
32	08182	5	20	5
33	08183	<5	5	5
34	08184	<5	5	5
35	08185	<5	5	5
36	08186	<5	<5	<5
37	08187	<5	<5	<5
38	08188	<5	<5	<5
39	08189	<5	<5	<5
40	08190	<5	<5	<5
41	08191	<5	<5	<5
42	08192	<5	<5	<5
43	08193	<5	5	<5
44	08194	<5	5	<5
45	08195	<5	<5	<5
46	08196	<5	<5	<5
47	08197	<5	5	<5
48	08198	5	<5	<5
49	08212	10	<5	<5
50	08213	<5	<5	<5
51	08214	50	35	5
52	08215	10	5	<5
53	08216	5	<5	<5
54	08217	10	<5	<5
55	08229	15	<5	<5
56	08230	5	<5	<5
57	08231	<5	<5	<5
58	08232	<5	<5	<5
59	08233	<5	<5	<5
60	08234	<5	<5	<5

QC DATA:

Resplit:

		Au	Pd	Pt
1	08151 Ace 02-01 13.0-14.5m	5	5	<5
36	08186	<5	<5	<5

Repeat:

		Au	Pd	Pt
1	08151 Ace 02-01 13.0-14.5m	<5	<5	5
10	08160 Ace 02-01 52.1-54.0m	<5	<5	<5
19	08169 Ace 02-01 88.0-89.5m	<5	<5	5
36	08186	<5	<5	<5
45	08195	5	5	<5
54	08217	5	<5	<5

Standard:

	Au	Pd	Pt
GEO'02	110	<5	5

JJ/kk
XLS/02
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CC: Wildrock Resources - Email & Mail

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

Eco Tech LABORATORY LTD.

8-Jul-02

ECO TECH LABORATORY LTD.
10041 Dallas Drive
KAMLOOPS, B.C.
V2C 6T4

Phone: 250-573-5700
Fax : 250-573-4557

ICP CERTIFICATE OF ANALYSIS AK 2002-124

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 60
Sample type: Core
Project #: ACE
Shipment #: None Given
Samples submitted by: Chris Wild

Values in ppm unless otherwise reported

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
1	08151 Ace 02-01 13.0-14.5m	0.2	1.02	<5	90	140	0.95	<1	18	108	57	3.75	40	0.87	345	2	0.01	26	720	48	<5	<20	18	0.13	<10	17	<10	15	58
2	08152 Ace 02-01 17.37-19.0m	<0.2	1.46	<5	45	<5	0.79	<1	21	78	22	3.92	40	1.11	245	<1	<0.01	33	910	4	<5	<20	20	0.09	<10	15	<10	12	76
3	08153 Ace 02-01 29.0-30.5m	0.4	1.97	<5	65	<5	1.80	1	21	65	34	4.48	30	1.18	347	2	0.01	40	790	6	<5	<20	34	0.07	<10	25	<10	12	124
4	08154 Ace 02-01 37.5-39.0m	0.2	1.07	<5	105	<5	3.10	4	14	85	75	3.37	10	0.81	578	16	0.02	50	1070	10	<5	<20	111	0.06	<10	40	<10	5	278
5	08155 Ace 02-01 45.0-46.5m	0.2	2.16	<5	35	<5	0.97	<1	22	66	19	4.54	30	1.17	339	<1	0.02	32	470	4	<5	<20	37	0.07	<10	14	<10	7	113
6	08156 Ace 02-01 47.5-49.0m	0.2	1.26	<5	80	<5	1.14	11	16	80	55	3.44	30	0.78	216	15	0.02	52	700	8	<5	<20	28	0.05	<10	53	<10	10	481
7	08157 Ace 02-01 49.0-49.7m	0.2	0.91	<5	85	<5	5.74	2	16	68	77	3.31	30	0.70	361	14	0.03	57	540	40	<5	<20	123	0.06	<10	35	<10	10	134
8	08158	0.2	0.46	<5	35	<5	>10	<1	4	17	9	1.05	10	0.52	317	2	0.02	86	340	36	<5	<20	2536	0.02	<10	10	<10	5	65
9	08159 Ace 02-01 50.7-52.1m	0.2	1.09	<5	125	<5	2.88	20	13	99	64	2.75	30	0.65	246	24	0.01	63	1200	10	<5	<20	116	0.05	<10	67	<10	8	682
10	08160 Ace 02-01 52.1-54.0m	0.2	0.59	<5	35	<5	>10	<1	5	18	7	1.24	10	0.63	236	<1	0.02	83	510	40	<5	<20	2748	0.03	<10	10	<10	5	43
11	08161 Ace 02-01 54.0-55.2m	<0.2	1.15	5	60	<5	1.95	1	19	89	37	4.03	20	0.97	214	7	0.02	39	470	24	<5	<20	112	0.06	<10	14	<10	5	246
12	08162 Ace 02-01 55.2-57.0m	<0.2	0.48	<5	105	<5	2.14	<1	20	89	80	3.53	10	0.64	289	18	0.02	58	1130	8	<5	<20	112	0.06	<10	28	<10	6	69
13	08163 Ace 02-01 57.0-58.8m	0.4	0.48	<5	90	<5	3.58	3	16	114	78	3.23	10	0.75	635	19	0.02	57	920	14	<5	<20	187	0.06	<10	28	<10	8	242
14	08164 Ace 02-01 58.8-61.0m	0.4	0.52	<5	85	<5	2.75	3	17	139	86	4.10	20	0.85	704	33	0.05	68	1160	42	<5	<20	124	0.08	<10	119	<10	9	181
15	08165 Ace 02-01 61.0-63.0m	0.4	0.68	<5	140	<5	3.82	2	16	108	81	4.08	20	1.17	1281	9	0.05	60	1130	42	<5	<20	151	0.10	<10	155	<10	11	212
16	08166 Ace 02-01 63.0-65.3m	0.2	0.70	<5	160	<5	3.44	3	17	102	69	3.86	30	0.85	640	25	0.04	51	1130	24	<5	<20	154	0.08	<10	59	<10	12	209
17	08167 Ace 02-01 65.3-67.5m	0.2	0.81	<5	100	<5	3.38	2	17	117	102	5.40	20	0.93	2065	11	0.02	64	1500	8	<5	<20	162	0.12	<10	37	<10	9	169
18	08168 Ace 02-01 76.0-77.5m	<0.2	2.34	<5	105	5	0.66	<1	30	171	28	5.48	40	1.29	153	<1	0.01	43	1180	6	<5	<20	24	0.11	<10	38	<10	16	127
19	08169 Ace 02-01 88.0-89.5m	<0.2	2.44	<5	55	<5	0.48	<1	33	122	38	5.91	40	1.31	123	2	0.01	50	890	6	<5	<20	18	0.10	<10	35	<10	14	134
20	08170 Ace 02-01 103.6-105.1m	0.2	2.15	<5	50	<5	1.20	<1	24	104	36	5.12	40	1.38	263	1	0.02	45	890	6	<5	<20	12	0.09	<10	39	<10	13	111
21	08171 Ace 02-01 130.0-131.5m	0.2	1.95	<5	25	<5	0.86	<1	28	89	51	5.76	30	1.12	512	2	<0.01	44	550	6	<5	<20	33	0.09	<10	18	<10	5	141
22	08172 Ace 02-01 139.5-141.0m	<0.2	0.87	<5	20	<5	0.91	<1	10	141	17	2.43	20	0.69	494	2	0.02	22	290	6	<5	<20	11	0.05	<10	13	<10	6	41
23	08173	<0.2	2.02	<5	95	<5	0.53	<1	30	121	57	5.14	40	1.13	89	2	0.02	40	980	6	<5	<20	6	0.13	<10	41	<10	16	121
24	08174	0.6	0.46	<5	30	<5	1.45	1	21	126	438	6.68	30	0.62	351	2	0.01	56	540	4	<5	<20	16	0.10	<10	4	<10	8	17
25	08175	<0.2	0.52	<5	30	<5	6.17	<1	16	101	33	3.72	20	1.95	1067	10	0.01	59	1460	6	<5	<20	64	0.08	<10	16	<10	12	34

8-Jul-02

ECO TECH LABORATORY LTD.
10041 Dallas Drive
KAMLOOPS, B.C.
V2C 6T4

Phone: 250-573-5700
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ICP CERTIFICATE OF ANALYSIS AK 2002-124B

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 5
Sample type: Core
Project #: Frank Creek
Shipment #: None Given
Samples submitted by: Chris Wild

Values in ppm unless otherwise reported

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
61	08235	0.2	0.17	35	30	<5	0.52	<1	16	107	60	2.60	10	0.34	137	4	<0.01	31	530	118	<5	<20	9	0.04	<10	7	<10	4	109
62	08236	0.6	0.40	80	70	<5	4.76	<1	23	78	79	3.71	10	2.65	1347	2	0.02	64	1090	162	<5	<20	68	0.08	<10	13	<10	6	234
63	08237	1.2	0.20	65	40	<5	6.19	3	17	63	125	3.69	10	3.12	2346	2	0.02	47	710	396	<5	<20	60	0.10	<10	4	10	6	994
64	08238	2.0	0.29	60	45	<5	0.64	5	26	86	93	3.56	10	0.69	674	<1	0.02	52	430	1374	<5	<20	11	0.06	<10	3	10	3	1725
65	08239	1.0	0.20	50	35	<5	2.10	3	17	97	80	2.72	10	1.17	1441	3	0.01	35	580	546	<5	<20	27	0.06	<10	3	<10	5	1085
QC DATA:																													
Standard:																													
GEO '02		1.6	1.59	50	140	<5	1.67	<1	21	64	81	3.81	10	0.94	623	<1	0.02	27	650	22	<5	<20	37	0.14	<10	73	<10	11	79

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

JJ/kk
df/
XLS/02
E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

BARKER MINERALS LTD. ICP CERTIFICATE OF ANALYSIS AK 2002-124 ECO TECH LABORATORY LTD.

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
26	08176	<0.2	2.20	<5	40	<5	0.29	<1	26	107	23	5.24	40	1.24	138	<1	0.02	44	430	8	<5	<20	8	0.09	<10	28	<10	9	134
27	08177	<0.2	0.68	<5	10	<5	0.40	<1	6	166	10	1.84	10	0.43	218	6	0.01	13	230	8	<5	<20	15	0.03	<10	10	<10	3	32
28	08178	0.2	0.97	<5	10	<5	0.48	<1	24	184	47	4.35	20	0.70	181	2	0.01	58	250	10	<5	<20	21	0.07	<10	14	<10	5	50
29	08179	<0.2	1.30	<5	15	<5	0.41	<1	14	154	17	3.12	20	0.86	423	4	0.02	30	320	8	<5	<20	14	0.05	<10	18	<10	6	60
30	08180	1.6	1.39	<5	20	<5	3.87	2	41	175	557	>10	40	1.57	7534	4	0.01	165	4100	20	<5	<20	252	0.38	<10	449	<10	30	282
31	08181	0.9	2.15	<5	25	5	5.63	2	38	187	340	>10	50	2.62	>10000	9	<0.01	127	3510	<2	<5	<20	291	0.75	40	546	<10	30	371
32	08182	0.4	1.65	<5	30	<5	4.91	4	31	159	393	>10	30	2.43	>10000	3	0.03	106	2260	<2	<5	<20	260	0.54	10	574	<10	24	534
33	08183	0.2	0.57	<5	60	<5	1.86	3	18	134	116	5.00	10	0.77	2488	19	0.02	51	1270	14	<5	<20	80	0.11	<10	17	<10	8	203
34	08184	0.2	0.75	<5	40	<5	3.02	2	24	111	115	7.06	20	1.22	3129	14	0.03	68	1580	28	<5	<20	164	0.16	<10	110	<10	13	214
35	08185	0.2	0.37	<5	85	<5	3.83	<1	18	93	30	3.90	10	1.05	570	26	0.07	50	1140	26	<5	<20	215	0.07	<10	54	<10	9	171
36	08186	0.4	0.33	<5	60	<5	3.61	1	15	86	35	3.61	10	1.19	986	40	0.06	53	1280	56	<5	<20	243	0.08	<10	59	<10	11	187
37	08187	0.4	0.26	<5	80	<5	2.65	1	12	95	70	3.48	10	0.72	1016	125	0.05	53	1120	32	<5	<20	127	0.07	<10	10	<10	8	51
38	08188	1.1	0.13	<5	70	5	1.73	1	10	140	35	2.29	10	0.47	578	163	0.06	48	450	108	<5	<20	106	0.04	<10	10	<10	4	99
39	08189	0.4	0.78	<5	50	<5	0.77	<1	22	97	56	5.21	20	1.23	387	1	0.08	33	390	10	<5	<20	56	0.10	<10	104	<10	4	139
40	08190	<0.2	1.49	<5	30	<5	0.71	<1	24	77	31	4.20	30	0.92	413	<1	<0.01	40	440	<2	<5	<20	29	0.08	<10	12	<10	5	104
41	08191	<0.2	0.34	<5	70	<5	3.54	<1	16	91	56	3.46	10	1.15	1121	8	<0.01	43	1000	2	<5	<20	111	0.07	<10	8	<10	6	84
42	08192	0.2	1.04	<5	90	<5	0.57	<1	25	87	54	4.97	20	1.06	268	<1	0.03	39	430	18	<5	<20	41	0.10	<10	67	<10	4	174
43	08193	0.2	1.45	<5	65	<5	0.68	<1	29	76	69	4.95	20	1.17	368	1	0.05	46	390	8	<5	<20	67	0.21	<10	138	<10	10	137
44	08194	<0.2	1.68	<5	115	10	0.19	<1	26	71	32	4.34	20	0.93	189	<1	<0.01	36	540	4	<5	<20	16	0.19	<10	22	<10	7	108
45	08195	0.2	0.95	<5	70	<5	0.69	1	28	71	64	5.90	20	1.14	381	<1	0.04	41	540	18	<5	<20	53	0.12	<10	61	<10	5	126
46	08196	<0.2	2.67	<5	30	<5	1.15	<1	28	111	43	5.93	50	1.56	566	3	0.01	53	720	6	<5	<20	46	0.09	<10	30	<10	9	112
47	08197	<0.2	2.23	<5	30	<5	0.47	<1	23	95	30	5.10	20	1.24	338	<1	0.02	39	560	8	<5	<20	27	0.07	<10	18	<10	3	100
48	08198	<0.2	2.48	<5	20	<5	0.81	<1	30	94	42	6.48	30	1.33	714	1	0.02	45	650	4	<5	<20	34	0.10	<10	26	<10	4	99
49	08212	0.2	0.20	15	40	<5	1.94	7	16	79	35	4.09	10	0.82	511	92	0.07	52	540	66	<5	<20	230	0.07	<10	36	<10	5	665
50	08213	0.2	0.15	10	55	<5	2.10	4	15	96	23	3.82	10	0.72	581	63	0.06	52	800	24	<5	<20	174	0.06	<10	26	<10	6	353
51	08214	1.6	0.43	70	25	<5	2.98	<1	63	133	671	>10	60	1.17	5747	10	<0.01	171	5500	<2	<5	<20	134	0.41	20	44	<10	16	217
52	08215	0.8	1.58	<5	80	<5	4.52	<1	25	117	159	>10	40	1.69	>10000	8	<0.01	55	1590	<2	<5	<20	219	0.39	10	96	<10	7	270
53	08216	<0.2	0.50	<5	155	<5	1.58	<1	12	149	53	3.55	20	0.49	3076	5	0.01	16	730	<2	<5	<20	71	0.10	<10	4	<10	5	77
54	08217	0.6	0.33	25	60	10	2.34	1	22	99	255	5.73	20	0.72	1515	67	0.04	69	1710	26	<5	<20	81	0.11	<10	19	<10	11	112
55	08229	0.2	2.29	<5	40	<5	0.46	<1	26	96	25	5.49	40	1.14	437	1	0.02	44	520	16	<5	<20	15	0.08	<10	20	<10	10	112
56	08230	0.2	0.39	<5	50	<5	1.91	11	17	94	66	4.21	20	0.72	667	45	0.04	60	1380	34	<5	<20	102	0.08	<10	42	<10	9	385
57	08231	0.2	0.53	<5	80	<5	4.34	1	28	97	122	5.83	30	1.37	2168	15	0.03	66	1760	22	<5	<20	191	0.13	<10	93	<10	13	159
58	08232	<0.2	0.58	<5	10	<5	1.07	<1	8	153	17	2.50	90	0.54	504	7	<0.01	11	350	<2	<5	<20	27	0.05	<10	7	<10	7	48
59	08233	0.2	0.68	<5	50	<5	0.88	<1	22	83	50	5.20	20	0.72	468	<1	0.04	29	340	6	<5	<20	52	0.08	<10	15	<10	4	57
60	08234	0.4	0.79	<5	50	<5	1.19	<1	28	81	60	6.07	20	1.11	584	<1	0.05	48	630	20	<5	<20	93	0.10	<10	65	<10	6	157

BARKER MINERALS LTD. ICP CERTIFICATE OF ANALYSIS AK 2002-124 ECO TECH LABORATORY LTD.

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
QC DATA:																													
Resplit:																													
1	08151 Ace 02-01 13.0-14.5m	0.2	1.08	<5	115	75	1.01	<1	21	77	54	4.07	40	0.85	360	<1	0.01	27	820	50	<5	<20	17	0.14	<10	19	<10	16	58
36	08186	0.4	0.35	<5	60	<5	4.01	1	17	87	34	3.91	10	1.18	1062	49	0.07	58	1390	62	<5	<20	234	0.08	<10	65	<10	13	199
Repeat:																													
1	08151 Ace 02-01 13.0-14.5m	0.2	1.01	<5	90	90	0.97	<1	19	110	55	3.78	40	0.86	350	2	0.01	27	720	40	<5	<20	18	0.13	<10	17	<10	15	59
10	08160 Ace 02-01 52.1-54.0m	0.2	0.58	<5	30	<5	>10	<1	5	20	6	1.29	10	0.60	246	<1	0.02	85	520	42	<5	<20	2564	0.03	<10	10	<10	5	46
19	08169 Ace 02-01 88.0-89.5m	<0.2	2.42	<5	55	<5	0.47	<1	33	123	41	5.93	40	1.30	126	2	0.01	51	880	8	<5	<20	17	0.10	<10	35	<10	14	136
36	08186	0.4	0.33	<5	70	<5	3.77	<1	16	87	35	3.72	10	1.20	1028	43	0.06	56	1340	62	<5	<20	243	0.08	<10	61	<10	12	194
45	08195	0.2	0.97	<5	95	<5	0.72	<1	28	76	63	5.99	20	1.16	405	<1	0.04	42	560	14	<5	<20	52	0.12	<10	63	<10	5	130
54	08217	0.6	0.33	25	65	10	2.30	1	22	99	252	6.15	20	0.70	1485	76	0.04	77	1660	26	<5	<20	80	0.11	<10	19	<10	11	113
Standard:																													
GEO '02		1.4	1.61	65	140	<5	1.66	<1	21	64	80	3.82	10	0.94	625	<1	0.02	28	660	20	5	<20	38	0.15	<10	74	<10	11	79
GEO '02		1.2	1.59	50	140	<5	1.67	<1	21	64	81	3.81	10	0.94	623	<1	0.02	27	650	22	<5	<20	37	0.14	<10	73	<10	11	79

JJ/kk
df/122a/124a/124b
XLS/02
E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

10-Jul-02

ECO TECH LABORATORY LTD.
10041 Dallas Drive
KAMLOOPS, B.C.
V2C 6T4

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ICP CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 24
Sample Type: Core
Project #: ACE
Drill Hole #: ACE - 02-04
Samples submitted by: Chris Wild

Values in ppm unless otherwise reported

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
1	8199	<0.2	1.40	<5	110	<5	0.88	<1	19	75	18	3.21	40	0.80	247	<1	0.02	28	860	4	<5	<20	31	0.13	<10	21	<10	15	66
2	8200	<0.2	1.64	<5	80	<5	0.75	<1	27	69	32	4.66	30	1.00	304	<1	0.02	38	640	6	<5	<20	37	0.11	<10	20	<10	8	95
3	8201	<0.2	1.16	<5	35	<5	7.37	<1	16	43	30	3.45	20	1.01	624	<1	0.01	50	1040	22	<5	<20	316	0.06	<10	9	<10	10	86
4	8202	0.2	1.10	<5	50	<5	2.16	<1	21	91	30	4.41	30	1.27	277	<1	0.05	41	610	26	<5	<20	145	0.10	<10	54	<10	8	126
5	8203	0.2	0.49	5	75	<5	>10	<1	11	78	20	2.39	20	1.03	389	26	0.05	62	650	76	<5	<20	574	0.06	<10	44	<10	13	153
6	8204	0.6	0.25	40	75	<5	9.51	<1	10	50	14	2.09	10	0.55	420	92	0.03	66	490	100	<5	<20	514	0.04	<10	25	<10	11	87
7	8205	0.2	0.47	15	65	<5	4.00	1	18	91	25	3.72	20	0.89	379	24	0.04	49	710	28	<5	<20	225	0.06	<10	21	<10	11	117
8	8206	0.2	0.40	90	75	<5	7.62	2	14	66	24	3.96	20	0.97	1283	12	0.02	60	1070	64	<5	<20	481	0.08	<10	20	<10	9	292
9	8207	0.2	0.31	860	35	<5	4.71	<1	22	177	114	>10	40	1.45	7915	10	<0.01	78	770	22	<5	<20	261	0.30	<10	20	<10	9	71
10	8208	5.2	0.26	60	45	10	2.78	9	16	88	31	3.84	20	0.72	576	44	0.05	77	2020	704	<5	<20	159	0.06	<10	27	<10	13	855
11	8209	0.2	0.38	90	25	<5	1.08	8	19	102	114	3.74	10	0.44	256	21	0.01	66	500	10	<5	<20	72	0.05	<10	31	10	5	767
12	8210	0.6	0.13	30	30	<5	1.70	6	15	138	24	4.28	20	0.47	507	230	0.05	74	1740	94	<5	<20	127	0.06	<10	16	<10	9	507
13	8211	2.0	0.21	5	30	<5	2.06	6	17	91	26	4.39	20	0.71	585	307	0.08	65	1020	312	<5	<20	189	0.06	<10	27	<10	7	485
14	8218	0.4	0.23	<5	65	<5	4.56	<1	20	63	69	4.21	20	0.97	982	36	0.04	67	1390	36	<5	<20	136	0.07	<10	19	<10	10	55
15	8219	0.3	0.88	<5	60	<5	2.86	<1	21	106	67	4.48	30	1.62	590	4	0.04	52	1150	26	<5	<20	91	0.07	<10	52	<10	8	152
16	8220	0.2	1.56	<5	65	<5	>10	<1	18	98	23	3.35	20	2.19	766	1	0.02	73	430	18	<5	<20	369	0.06	<10	15	<10	11	57
17	8221	0.2	0.35	10	60	<5	8.06	<1	20	73	99	5.21	30	2.26	2386	25	0.03	84	1070	24	<5	<20	249	0.11	<10	18	<10	12	68
18	8222	0.4	0.33	35	50	<5	3.72	1	29	92	138	5.05	40	1.05	1023	39	0.05	68	1850	28	<5	<20	122	0.09	<10	32	<10	12	108
19	8223	0.6	0.28	<5	90	<5	6.16	2	13	81	76	3.93	30	1.06	2108	8	0.02	50	1550	32	<5	<20	165	0.09	<10	31	<10	14	121
20	8224	0.8	0.31	<5	45	<5	3.19	2	16	126	34	3.63	30	0.90	707	95	0.07	72	1750	94	<5	<20	119	0.07	<10	38	<10	11	187
21	8225	0.2	0.75	<5	50	<5	4.12	4	14	134	47	3.20	30	1.00	622	29	0.08	70	3700	50	<5	<20	174	0.12	<10	166	<10	22	179
22	8226	0.2	1.36	<5	65	<5	1.03	1	24	88	72	5.57	40	1.20	384	15	0.03	49	800	10	5	<20	34	0.07	<10	51	<10	7	155
23	8227	<0.2	1.83	<5	55	<5	0.72	<1	23	89	32	5.09	30	1.12	302	1	0.02	45	610	10	<5	<20	35	0.06	<10	18	<10	5	139
24	8228	0.2	1.05	<5	80	<5	3.63	1	24	83	50	4.55	40	1.01	732	11	0.05	54	780	22	<5	<20	158	0.08	<10	52	<10	12	130

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
<u>QC DATA:</u>																													
Resplit:																													
1	8199	<0.2	1.46	<5	135	<5	0.97	<1	21	95	17	3.51	40	0.81	269	<1	0.03	32	940	8	5	<20	30	0.15	<10	23	<10	17	70
Repeat:																													
1	8199	<0.2	1.35	<5	115	<5	0.93	<1	19	80	17	3.36	40	0.77	260	<1	0.02	31	900	6	<5	<20	29	0.13	<10	22	<10	16	71
10	8208	5.2	0.27	65	45	5	2.76	9	16	89	32	3.85	20	0.74	570	45	0.05	74	2000	712	<5	<20	163	0.06	<10	28	<10	13	851
19	8223	0.6	0.25	<5	80	<5	5.50	1	12	75	66	3.74	30	0.92	1872	6	0.02	45	1410	28	<5	<20	140	0.09	<10	28	<10	13	110
Standard:																													
GEO '02		1.6	1.80	65	160	<5	1.83	<1	23	75	90	4.23	20	1.05	684	<1	0.03	36	720	22	15	<20	42	0.17	<10	84	<10	12	79

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

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10-Jul-02

ECO TECH LABORATORY LTD.
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ICP CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 65
Sample Type: Core
Project #: Frank Creek
Drill Hole #: FC-02-01 to FC-02-06
Samples submitted by: Chris Wild

Values in ppm unless otherwise reported

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
25	8240	1.8	0.24	75	45	<5	0.30	5	12	68	88	4.24	20	1.26	1395	13	0.02	57	560	618	<5	<20	11	0.07	<10	17	<10	3	2031
26	8241	18.0	0.25	130	40	<5	0.52	64	31	88	1183	>10	40	2.46	2498	8	0.02	57	530	3226	<5	<20	19	0.18	<10	19	<10	3	>10000
27	8242	1.6	0.22	50	45	<5	3.72	9	18	59	217	4.65	20	2.45	2218	<1	0.02	42	260	208	<5	<20	51	0.10	<10	3	<10	5	2061
28	8243	12.4	0.40	115	55	<5	0.30	127	68	130	1481	>10	50	3.55	3957	<1	0.03	71	360	808	<5	<20	8	0.27	<10	15	<10	4	>10000
29	8244	0.6	0.59	65	100	<5	0.47	3	28	143	57	>10	50	2.62	3100	8	0.03	57	1450	60	<5	<20	19	0.17	<10	25	<10	6	622
30	8245	0.6	0.34	5	40	<5	0.28	5	10	149	40	3.24	30	1.51	718	5	0.03	22	110	502	<5	<20	14	0.05	<10	3	<10	2	1379
31	8246	0.2	0.33	<5	45	<5	1.50	<1	10	128	14	2.81	20	0.81	1423	5	0.02	25	200	6	<5	<20	31	0.06	<10	3	<10	2	45
32	8247	<0.2	0.87	40	55	<5	>10	<1	51	120	64	7.27	40	5.09	1184	<1	0.06	214	1730	4	<5	<20	107	0.12	<10	50	<10	5	71
33	8248	<0.2	0.47	145	235	<5	>10	<1	48	61	35	7.64	40	3.08	1011	2	0.04	167	2010	6	<5	<20	114	0.12	<10	19	<10	10	57
34	8249	<0.2	0.27	<5	65	<5	5.58	<1	12	110	84	3.47	10	1.86	942	5	0.01	51	160	8	<5	<20	189	0.06	<10	4	<10	6	46
35	8250	<0.2	0.37	150	145	<5	>10	<1	55	95	70	7.01	30	5.49	1066	<1	0.05	291	1440	4	<5	<20	139	0.11	<10	33	<10	9	67
36	8251	0.2	0.29	<5	35	<5	0.18	1	14	154	108	6.44	20	0.38	22	3	0.01	36	200	<2	<5	<20	8	0.07	<10	4	<10	2	43
37	8252	0.2	0.21	<5	50	<5	1.94	<1	6	102	10	1.66	10	0.78	534	2	0.03	20	190	20	<5	<20	61	0.03	<10	2	<10	2	24
38	8253	<0.2	1.36	150	15	<5	>10	<1	57	201	76	6.60	10	4.33	1229	<1	0.04	233	920	12	<5	<20	238	0.10	<10	31	<10	6	79
39	8254	0.4	0.29	<5	35	<5	0.11	<1	10	116	11	2.36	20	0.49	246	2	0.02	19	240	14	<5	<20	4	0.03	<10	2	<10	3	45
40	8255	>30	0.13	8010	15	<5	0.10	<1	46	151	5238	>10	50	0.45	7	8	0.01	53	290	2866	65	<20	4	0.29	<10	16	<10	3	3333
41	8256	0.8	3.68	<5	225	<5	5.43	<1	30	229	48	6.04	80	4.38	762	<1	0.02	200	2390	20	<5	<20	96	0.10	<10	92	<10	9	76
42	8257	0.4	0.48	20	35	<5	0.65	<1	15	88	16	3.88	20	1.66	278	<1	0.04	30	580	20	<5	<20	16	0.05	<10	3	<10	4	77
43	8258	0.2	1.80	475	35	<5	9.57	<1	63	434	8	7.56	30	6.78	1819	<1	0.03	392	820	6	<5	<20	333	0.13	<10	57	<10	6	137
44	8259	0.2	2.54	65	20	<5	7.83	<1	50	306	102	7.75	30	4.82	1426	2	0.04	142	1040	8	<5	<20	83	0.13	<10	121	<10	4	96
45	8260	0.2	0.80	<5	55	<5	0.66	<1	16	131	15	3.89	40	1.08	425	<1	0.04	34	600	18	<5	<20	18	0.05	<10	6	<10	5	91
46	8261	<0.2	1.42	40	55	<5	7.83	<1	46	98	108	7.49	30	3.27	1452	<1	0.06	96	1120	<2	<5	<20	129	0.12	<10	52	<10	4	86
47	8262	<0.2	0.75	500	40	<5	>10	<1	83	450	57	6.10	30	6.25	2608	<1	0.02	788	370	4	<5	<20	410	0.14	<10	23	<10	4	72
48	8263	<0.2	0.27	<5	155	<5	3.15	<1	10	108	10	2.80	20	1.16	649	2	0.03	32	550	<2	<5	<20	92	0.05	<10	3	<10	6	43
49	8264	0.2	0.16	<5	220	<5	1.28	4	5	197	33	1.31	<10	0.46	306	9	0.01	27	300	24	<5	<20	54	0.02	<10	7	<10	2	371

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
50	8265	<0.2	0.29	30	55	<5	7.42	<1	24	101	15	5.03	30	1.60	976	<1	0.03	84	830	10	<5	<20	128	0.08	<10	5	<10	6	71
51	8266	6.2	0.30	175	20	<5	0.19	8	35	130	1970	>10	30	1.69	1314	3	0.03	28	500	280	<5	<20	6	0.18	<10	8	<10	3	2268
52	8267	5.6	0.97	335	20	<5	0.30	3	53	136	2751	>10	30	1.77	1261	2	0.04	37	1160	430	<5	<20	8	0.21	<10	14	<10	4	1252
53	8268	5.2	0.80	210	25	<5	0.48	2	33	159	807	>10	30	2.17	1466	3	0.06	36	1880	232	<5	<20	14	0.17	<10	11	<10	6	872
54	8269	4.8	1.37	220	30	<5	0.10	3	28	118	908	>10	30	1.89	1212	1	0.03	29	210	486	<5	<20	5	0.15	<10	17	<10	3	1349
55	8270	8.6	0.20	165	20	<5	0.09	5	48	152	4935	>10	30	1.48	1230	6	0.03	37	320	260	<5	<20	4	0.25	<10	7	<10	3	1255
56	8271	3.4	0.36	50	25	<5	0.09	2	24	165	539	8.62	30	1.43	1091	1	0.04	28	120	164	<5	<20	4	0.12	<10	6	<10	3	502
57	8272	1.2	0.39	45	35	<5	0.11	2	28	141	55	>10	40	1.83	1678	2	0.05	39	60	56	<5	<20	5	0.15	<10	9	<10	4	768
58	8273	10.5	0.40	125	30	<5	0.15	9	77	131	2104	>10	50	2.95	2448	<1	0.05	37	20	784	<5	<20	5	0.33	<10	6	<10	4	1671
59	8274	2.2	1.13	25	30	<5	0.12	3	25	119	342	>10	30	2.04	1553	2	0.03	26	170	156	<5	<20	5	0.14	<10	13	<10	2	750
60	8275	2.8	0.90	35	25	<5	0.15	2	24	129	1173	9.64	30	1.94	1561	<1	0.03	23	330	120	<5	<20	5	0.16	<10	10	<10	3	501
61	8276	2.0	2.34	80	20	<5	0.48	5	41	138	1514	>10	30	2.52	1559	2	0.02	34	1480	200	<5	<20	12	0.20	<10	28	<10	5	1380
62	8277	3.4	2.43	695	20	<5	0.37	3	34	149	2315	>10	30	2.60	1529	3	0.01	35	880	232	<5	<20	9	0.19	<10	29	<10	3	1926
63	8278	4.6	2.86	180	20	<5	0.12	13	71	162	3612	>10	50	3.13	1780	2	0.02	39	270	234	<5	<20	4	0.32	<10	27	<10	4	4145
64	8279	5.3	2.04	195	20	<5	0.14	14	109	142	3189	>10	70	3.29	2924	4	0.01	57	120	296	<5	<20	4	0.40	<10	20	<10	4	3522
65	8280	3.0	0.89	135	35	<5	0.17	3	39	138	861	>10	30	2.31	1583	1	0.03	40	250	160	<5	<20	5	0.17	<10	8	<10	3	821
66	8281	0.6	1.28	50	40	<5	0.13	<1	13	129	94	6.50	20	1.62	1208	2	0.03	38	320	44	<5	<20	5	0.10	<10	10	<10	3	233
67	8282	0.2	0.27	<5	55	<5	3.62	<1	9	105	130	3.63	20	1.49	1656	2	0.03	28	100	54	<5	<20	36	0.08	<10	2	<10	4	124
68	8283	0.2	0.75	25	50	<5	1.57	<1	25	123	48	5.53	30	1.64	871	<1	0.04	54	530	52	<5	<20	25	0.08	<10	6	<10	4	256
69	8284	0.2	1.69	20	45	<5	0.17	<1	19	99	30	4.55	40	1.45	710	1	0.02	40	600	10	<5	<20	7	0.06	<10	10	<10	5	152
70	8285	<0.2	1.04	45	20	<5	7.66	<1	41	164	42	6.86	30	4.41	1378	<1	0.06	154	1080	4	<5	<20	38	0.11	<10	54	<10	5	93
71	8286	0.6	0.51	320	50	<5	0.10	<1	17	113	24	6.54	20	1.67	1144	1	0.04	32	150	326	<5	<20	6	0.09	<10	5	<10	2	693
72	8287	0.6	1.85	370	40	<5	0.19	<1	23	77	48	8.35	20	2.08	1192	<1	0.02	40	630	270	<5	<20	10	0.12	<10	17	<10	4	783
73	8288	0.2	0.82	165	45	<5	0.12	<1	18	99	20	5.78	20	1.48	927	<1	0.03	30	270	76	<5	<20	8	0.08	<10	9	<10	3	606
74	8289	0.8	1.89	65	35	<5	0.26	4	23	117	64	7.98	30	1.86	1090	3	0.02	39	590	166	10	<20	11	0.11	<10	16	<10	4	943
75	8290	0.6	1.76	25	45	<5	0.13	1	19	112	27	6.56	30	1.62	860	<1	0.02	36	300	192	<5	<20	7	0.09	<10	12	<10	3	378
76	8291	1.0	1.84	20	35	<5	0.16	3	21	106	64	8.24	20	1.93	1167	<1	0.02	33	140	290	<5	<20	6	0.11	<10	14	<10	3	931
77	8292	0.6	1.90	30	45	<5	0.20	1	19	118	28	7.55	30	1.81	1175	1	0.02	33	140	170	<5	<20	7	0.11	<10	14	<10	3	335
78	8293	0.6	1.99	250	40	<5	0.22	<1	20	125	40	7.54	20	1.90	1291	1	0.02	35	290	192	<5	<20	9	0.11	<10	16	<10	3	614
79	8294	1.0	1.98	105	55	<5	0.06	3	18	86	39	6.05	30	1.53	934	1	0.02	34	130	370	<5	<20	4	0.09	<10	20	<10	2	969
80	8295	4.2	1.77	50	40	<5	0.15	8	32	88	284	>10	30	2.34	1550	<1	0.01	37	220	1006	<5	<20	7	0.16	<10	21	<10	3	2635
81	8296	2.0	0.46	425	35	<5	0.23	5	28	98	204	>10	30	2.16	1435	<1	0.04	34	360	920	<5	<20	11	0.14	<10	10	<10	3	2234
82	8297	20.2	0.99	335	25	<5	2.10	6	61	109	3978	>10	30	3.55	2321	1	0.04	45	470	1292	<5	<20	43	0.29	<10	13	<10	4	2507
83	8298	0.4	2.68	80	60	<5	3.97	<1	33	146	48	7.22	30	4.81	3034	<1	0.04	95	1040	68	<5	<20	58	0.15	<10	58	<10	5	301
84	8299	1.4	0.56	55	55	<5	0.19	2	22	104	248	7.10	30	1.72	1228	1	0.04	34	200	100	<5	<20	8	0.11	<10	5	<10	3	603

BARKER MINERALS LTD.

ICP CERTIFICATE OF ANALYSIS AK 2002-131

ECO TECH LABORATORY LTD.

Et #.	Tag #	Ag	Al %	As	Ba	Bi	Ca %	Cd	Co	Cr	Cu	Fe %	La	Mg %	Mn	Mo	Na %	Ni	P	Pb	Sb	Sn	Sr	Ti %	U	V	W	Y	Zn
85	8300	2.4	0.28	40	40	<5	0.09	2	29	140	1066	9.35	30	0.91	943	3	0.04	19	80	26	<5	<20	6	0.15	<10	3	<10	3	225
86	8301	4.0	0.25	35	30	<5	0.08	2	63	140	2276	>10	30	1.01	949	2	0.04	21	100	112	<5	<20	5	0.20	<10	3	<10	3	317
87	8302	1.2	0.42	<5	60	<5	0.47	1	23	121	426	8.36	30	0.91	1050	<1	0.04	25	1710	80	<5	<20	15	0.12	<10	4	<10	6	359
88	8303	1.2	0.33	<5	65	<5	0.18	2	25	91	173	7.12	20	1.10	1181	1	0.03	37	160	152	<5	<20	6	0.10	<10	3	<10	3	443
89	8304	0.8	0.34	10	55	<5	1.73	4	26	104	175	5.96	20	1.24	1084	<1	0.03	32	320	76	<5	<20	30	0.09	<10	3	<10	4	776
Repeat:																													
36	8251	0.2	0.28	<5	40	<5	0.19	1	14	150	105	6.34	20	0.37	25	3	0.01	36	210	<2	<5	<20	8	0.07	<10	4	<10	2	42
45	8260	0.2	0.79	<5	55	<5	0.66	<1	16	130	16	3.86	40	1.06	425	<1	0.04	32	590	16	<5	<20	18	0.05	<10	6	<10	5	95
54	8269	4.6	1.36	225	25	<5	0.10	3	28	118	914	>10	30	1.87	1196	<1	0.03	29	200	478	<5	<20	5	0.15	<10	17	<10	3	1339
71	8286	0.6	0.51	330	50	<5	0.10	<1	16	115	24	6.53	20	1.66	1145	<1	0.04	30	170	322	<5	<20	6	0.09	<10	5	<10	3	703
80	8295	4.2	1.78	55	40	<5	0.15	9	31	88	283	>10	30	2.35	1557	<1	0.01	36	240	1004	<5	<20	7	0.16	<10	21	<10	3	2632
Resplit:																													
36	8251	0.2	0.32	<5	30	<5	0.19	<1	14	136	104	6.55	20	0.38	23	3	0.01	37	180	<2	<5	<20	8	0.08	<10	4	<10	2	43
71	8286	0.6	0.45	410	45	<5	0.10	<1	16	113	22	6.69	20	1.72	1196	<1	0.03	31	140	312	<5	<20	6	0.09	<10	4	<10	3	709
Standard:																													
GEO '02		1.6	1.64	55	145	<5	1.74	<1	21	71	81	3.96	20	0.95	634	<1	0.03	34	650	18	10	<20	40	0.15	<10	78	<10	11	81
GEO '02		1.6	1.58	50	140	<5	1.70	<1	21	70	78	3.85	10	0.91	622	<1	0.03	33	640	20	5	<20	38	0.15	<10	76	<10	10	79

JJ/kk
df/131
XLS/02
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CC: Wildrock Resources - Email & Mail

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

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10041 Dallas Drive, Kamloops, B.C. V2C 6T4
Phone (250) 573-5700 Fax (250) 573-4557
email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-124A

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

July 10, 2002

ATTENTION: Louis Doyle

No. of samples received: 60
Sample type: Core
Project #: ACE
Shipment #: None Given
Samples submitted by: Chris Wild

Note: Values expressed in percent

ET #.	Tag #	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
1	08151 Ace 02-01 13.0-14.5m	0.11	0.18	71.72	0.07	6.32	1.94	11.64	1.39	0.90	0.39	4.14	1.29
2	08152 Ace 02-01 17.37-19.0m	0.11	0.24	66.84	0.06	6.68	2.18	14.81	1.11	0.97	0.35	4.24	2.41
3	08153 Ace 02-01 29.0-30.5m	0.17	0.20	63.09	0.07	6.89	2.19	16.51	2.47	0.89	0.74	3.56	3.22
4	08154 Ace 02-01 37.5-39.0m	0.22	0.25	65.31	0.09	5.07	1.55	12.99	3.93	0.53	0.96	2.70	6.53
7	08157 Ace 02-01 49.0-49.7m	0.17	0.14	61.83	0.06	4.91	1.41	14.10	7.55	0.51	1.92	2.52	4.87
14	08164 Ace 02-01 58.8-61.0m	0.09	0.27	70.41	0.11	5.99	1.61	11.38	3.65	0.42	4.65	1.02	0.90
16	08166 Ace 02-01 63.0-65.3m	0.22	0.28	65.95	0.10	5.47	1.68	13.52	4.36	0.53	3.88	2.34	1.91
18	08168 Ace 02-01 76.0-77.5	0.17	0.33	60.99	0.12	9.21	2.34	17.17	1.40	1.23	1.39	3.08	2.57
20	08170 Ace 02-01 103.6-105.1m	0.10	0.26	58.55	0.08	7.81	2.58	19.46	2.15	0.85	1.73	3.58	2.85
21	08171 Ace 02-01 130.0-131.5m	0.08	0.17	59.81	0.10	8.70	2.22	18.76	1.12	1.04	0.39	4.59	3.08
22	08172 Ace 02-01 139.5-141.0m	0.03	0.08	81.55	0.07	3.29	1.30	7.67	1.25	0.42	1.25	1.28	1.83
23	08173	0.12	0.27	62.11	0.10	8.49	2.08	17.42	1.44	0.98	1.73	3.44	1.82
25	08175	0.04	0.36	59.16	0.15	5.49	3.29	9.95	7.31	0.66	0.27	2.86	10.46
26	08176	0.10	0.13	60.93	0.11	8.68	2.33	18.34	0.99	1.11	1.34	3.27	2.72
29	08179	0.02	0.11	80.46	0.07	4.26	1.53	8.70	0.69	0.47	1.06	1.21	1.48
32	08182	0.19	0.65	51.16	1.81	18.97	4.05	8.08	6.24	0.83	2.38	1.85	3.88
33	08183	0.38	0.32	71.59	0.34	6.77	1.64	10.12	2.29	0.36	1.74	2.18	2.33
35	08185	0.12	0.36	62.50	0.09	5.21	1.80	13.98	4.76	0.49	6.70	1.01	3.03
37	08187	0.18	0.32	70.14	0.14	4.82	1.24	11.50	3.34	0.39	4.07	1.49	2.43
39	08189	0.06	0.13	59.30	0.07	7.95	2.09	17.03	1.20	0.94	8.23	1.23	1.78
40	08190	0.08	0.14	60.24	0.09	8.20	2.15	18.53	1.01	1.13	0.45	5.44	2.59
41	08191	0.13	0.26	65.87	0.16	5.65	2.31	10.96	4.52	0.55	0.01	3.76	5.88
42	08192	0.13	0.18	60.77	0.05	7.69	2.27	17.97	0.86	0.79	4.56	3.24	1.50
45	08195	0.17	0.19	57.49	0.07	8.70	2.37	18.58	1.05	0.91	5.74	3.26	1.52
47	08197	0.09	0.19	60.60	0.06	7.40	2.26	19.37	0.69	0.83	1.36	3.81	3.38
48	08198	0.07	0.21	57.99	0.11	9.35	2.43	18.85	1.14	1.27	1.16	3.57	3.89
49	08212	0.13	0.18	64.77	0.08	5.70	1.43	14.78	2.54	0.44	7.52	0.78	1.70
50	08213	0.15	0.28	68.13	0.09	5.26	1.24	12.58	2.71	0.39	6.24	0.81	2.18

ET #.		Tag #	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
51	08214		0.11	1.58	37.85	0.93	34.15	2.12	4.10	4.51	0.18	0.01	1.22	13.34
53	08216		0.33	0.21	73.77	0.40	5.41	1.07	9.82	1.89	0.28	0.13	3.02	3.66
55	08229		0.11	0.12	60.79	0.08	7.83	2.15	19.37	0.80	0.95	1.52	3.56	2.72
56	08230		0.25	0.31	68.23	0.10	5.85	1.46	12.83	2.41	0.44	4.59	1.84	1.67
60	08234		0.08	0.15	52.55	0.08	7.53	1.83	15.78	1.49	0.85	6.46	1.64	11.56

QC DATA:

Standard:

	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
Mrg-1	0.01	0.01	40.03	0.18	17.27	13.01	8.46	14.24	3.52	0.69	0.40	2.22
Sy-2	0.03	0.13	50.58	0.12	6.30	0.74	20.71	7.87	0.28	7.03	1.65	4.56

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Jutta Jealouse
B.C. Certified Assayer

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CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

10-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 24
Sample Type: Core
Project #: ACE
Drill Hole #: ACE - 02-04
Samples submitted by: Chris Wild

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
1	8199	5	<5	<5
2	8200	5	<5	<5
3	8201	<5	<5	5
4	8202	5	<5	<5
5	8203	5	5	<5
6	8204	5	<5	<5
7	8205	5	<5	<5
8	8206	40	<5	<5
9	8207	575	<5	<5
10	8208	5	<5	<5
11	8209	10	<5	<5
12	8210	15	5	<5
13	8211	10	5	5
14	8218	5	5	<5
15	8219	<5	<5	<5
16	8220	5	<5	<5
17	8221	5	<5	<5
18	8222	15	5	<5
19	8223	5	<5	<5
20	8224	5	<5	<5
21	8225	5	10	<5
22	8226	5	5	<5
23	8227	5	<5	<5
24	8228	5	<5	<5

BARKER MINERALS LTD. AK 2002-131 10-Jul-02

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
QC DATA:				
Resplit:				
1	8199	5	<5	<5
Repeat:				
1	8199	5	<5	<5
9	8207	615	-	-
10	8208	10	<5	<5
19	8223	10	<5	<5
Standard:				
GEO'02		130	-	-

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Jutta Jealouse
B.C. Certified Assayer

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Eco Tech LABORATORY LTD.

ASSAYING
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ENVIRONMENTAL TESTING

10041 Dallas Drive, Kamloops, B.C. V2C 6T4
Phone (250) 573-5700 Fax (250) 573-4557
email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

10-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 65
Sample Type: Core
Project #: Frank Creek
Drill Hole #: FC-02-01 to FC-02-06
Samples submitted by: Chris Wild

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
25	8240	5	<5	<5
26	8241	15	5	<5
27	8242	5	5	<5
[illegible]	8243	20	5	<5
29	8244	5	<5	<5
30	8245	5	<5	<5
31	8246	<5	<5	<5
32	8247	35	5	<5
33	8248	5	<5	<5
34	8249	5	<5	<5
35	8250	5	<5	<5
36	8251	20	<5	<5
37	8252	5	<5	<5
38	8253	10	<5	<5
39	8254	5	<5	<5
40	8255	250	<5	<5
41	8256	5	<5	5
42	8257	5	<5	<5
43	8258	5	<5	<5
44	8259	5	<5	<5
45	8260	<5	<5	5
46	8261	5	<5	<5
47	8262	5	10	<5
48	8263	<5	<5	<5
49	8264	5	<5	<5
50	8265	<5	<5	5
51	8266	95	<5	<5
[illegible]	8267	25	<5	<5

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
53	8268	110	<5	<5
54	8269	15	<5	<5
55	8270	115	<5	<5
56	8271	25	<5	<5
57	8272	15	<5	<5
58	8273	50	<5	<5
59	8274	10	<5	<5
60	8275	15	<5	<5
61	8276	20	<5	<5
62	8277	15	<5	<5
63	8278	35	<5	<5
64	8279	<5	90	<5
65	8280	65	<5	<5
66	8281	10	<5	<5
67	8282	5	<5	<5
68	8283	5	<5	<5
69	8284	<5	<5	<5
70	8285	<5	<5	<5
71	8286	10	<5	<5
72	8287	10	<5	<5
73	8288	10	<5	<5
74	8289	5	<5	<5
[illegible]	8290	<5	<5	<5
76	8291	5	<5	<5
77	8292	<5	<5	<5
78	8293	<5	<5	<5
79	8294	<5	<5	<5
80	8295	5	<5	<5
81	8296	10	<5	<5
82	8297	40	<5	<5
83	8298	5	<5	<5
84	8299	5	<5	<5
85	8300	150	<5	<5
86	8301	110	5	<5
87	8302	15	<5	<5
88	8303	5	<5	<5
89	8304	5	<5	<5

ET #.	Tag #	Au (ppb)	Pd (ppb)	Pt (ppb)
QC DATA:				
Resplit:				
36	8251	25	<5	<5
71	8286	25	<5	<5
Repeat:				
36	8251	15	<5	<5
45	8260	<5	<5	<5
54	8269	15	<5	<5
71	8286	15	<5	<5
80	8295	5	<5	<5
Standard:				
GEO'02		130	-	-
GEO'02		115	-	-
GEO'02		120	-	-

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email: ecotech@direct.ca

CERTIFICATE OF ASSAY AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

12-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 65
Sample Type: Core
Project #: Frank Creek
Drill Hole #: FC-02-01 to FC-02-06
Samples submitted by: Chris Wild

ET #.	Tag #	Ag (g/t)	Ag (oz/t)	Zn (%)
26	8241	-	-	2.30
28	8243	-	-	3.40
40	8255	83.2	2.43	-

QC DATA:

Repeat:				
R26	8241	-	-	2.20
R40	8255	83.6	2.44	-

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ENVIRONMENTAL TESTING

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email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-124B

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

12-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 5
Sample type: Core
Project #: Frank Creek
Shipment #: None Given
Samples submitted by: Chris Wild

Note: Values expressed in percent

ET #.	Tag #	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
62	08236	0.13	0.30	59.04	0.21	5.95	5.24	15.48	6.59	0.56	0.34	4.21	1.95
64	08237	0.10	0.12	69.62	0.09	4.98	1.54	14.02	0.81	0.61	0.33	3.98	3.79

QC DATA:
Standard:

		BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
Mrg-1		0.01	0.05	40.03	0.18	17.27	13.01	8.46	14.24	3.52	0.69	0.40	2.22
Sy-2		0.03	0.13	50.58	0.12	6.30	0.74	20.71	7.87	0.28	7.03	1.65	4.56

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

df/wr124c
.S/02
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email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-131

17-Jul-02

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 24
Sample Type: Core
Project #: ACE
Drill Hole #: ACE - 02-04
Samples submitted by: Chris Wild

Note: Values expressed in percent

ET #.	Tag #	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
1	8199	0.12	0.23	68.20	0.06	5.93	1.94	13.57	1.47	1.02	1.75	3.43	2.29
2	8200	0.17	0.19	61.51	0.07	8.15	2.13	16.59	1.05	1.16	1.93	4.00	3.05
3	8201	0.12	0.21	52.50	0.09	5.88	1.05	15.27	9.61	0.59	0.33	4.62	9.74
4	8202	0.01	0.70	60.84	0.06	6.09	3.25	14.98	8.35	0.71	0.67	1.11	3.55
5	8203	0.06	0.12	56.56	0.06	3.44	1.70	8.59	14.18	0.34	3.96	0.39	10.62
6	8204	0.10	0.12	59.96	0.06	3.05	1.15	8.81	12.70	0.33	2.90	1.35	9.47
7	8205	0.18	0.13	58.70	0.06	5.83	1.87	13.34	5.41	0.66	3.96	2.35	7.51
8	8206	0.33	0.24	52.89	0.17	5.92	2.01	12.27	9.77	0.42	0.93	3.45	11.60
10	8208	0.13	0.44	66.21	0.08	5.09	1.30	11.07	3.55	0.39	5.04	0.63	6.04
12	8210	0.09	0.32	74.53	0.07	5.58	0.78	7.56	2.03	0.29	3.89	0.23	4.62
13	8211	0.16	0.14	62.69	0.08	5.96	1.19	13.41	2.46	0.45	7.14	0.52	5.79
14	8218	0.17	0.19	62.23	0.13	5.77	1.58	11.48	5.67	0.44	4.55	1.08	6.71
15	8219	0.13	0.16	64.62	0.08	6.19	2.77	12.20	3.53	0.73	3.21	1.71	4.66
16	8220	0.14	0.10	47.91	0.11	5.25	3.84	11.25	14.31	0.47	0.77	2.41	13.46
17	8221	0.14	0.15	54.19	0.32	8.29	3.68	9.67	10.50	0.69	2.30	1.83	8.22
18	8222	0.11	0.31	60.96	0.14	6.68	1.64	12.68	4.67	0.64	5.24	1.06	5.87
19	8223	0.15	0.27	65.23	0.28	5.87	1.93	7.09	7.80	0.37	1.78	1.09	8.14
20	8224	0.12	0.28	66.88	0.10	4.73	1.49	10.80	3.99	0.50	5.30	0.65	5.16
21	8225	0.17	0.73	64.19	0.09	4.18	1.56	12.11	5.25	0.46	6.18	0.70	4.38
22	8226	0.17	0.16	59.35	0.06	8.24	2.35	17.32	1.40	0.81	3.94	2.81	3.39
24	8228	0.19	0.24	57.82	0.11	6.79	1.90	16.21	4.74	0.81	4.96	2.07	4.14

<u>**QC DATA:**</u>

Standard:	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
Mrg-1	0.01	0.03	38.49	0.18	17.73	13.42	8.47	14.79	3.74	0.77	0.14	2.22
Sy-2	0.03	0.14	49.38	0.12	6.56	0.66	20.98	8.33	0.31	7.25	1.70	4.56

/wr131
XLS/02
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CC: Wildrock Resources - Email & Mail

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0220085
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0220085

SAMPLE	PREP CODE		Cu %	Zn %
1LF05-12	212	--	1.14	-----
2LF01-44	212	--	-----	3.02
3LF01-32	212	--	-----	3.42
3LF06-03	212	--	2.08	-----
3LF06-05	212	--	1.27	-----

CERTIFICATION:

OVERLIMITS from A0219395

ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-200
Invoice No. : I0219395
P.O. Number :
Account :PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Weight Kg	Au ppb ICP	Pt ppb ICP	Pd ppb ICP	Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %
A0200-51	94139402	0.62	< 2	< 5	< 2	14.37	0.12	1.19	0.01	5.67	3.50	1.66	0.08	2.18	0.19	67.01	0.01	1.03	2.18	99.20
A0201-04	94139402	0.48	764	< 5	2	18.68	0.17	1.19	0.01	10.09	3.89	2.22	0.07	2.37	0.15	56.38	0.03	1.19	3.29	99.73
A0201-12	94139402	0.90	6	< 5	< 2	18.62	0.10	1.20	0.01	7.20	3.43	2.22	0.09	1.38	0.12	60.46	0.02	0.83	3.60	99.28
A0201-13	94139402	0.76	8	< 5	< 2	3.40	0.01	1.04	0.03	13.45	1.03	0.74	0.08	0.07	0.04	74.28	< 0.01	0.37	3.85	98.39
A0201-14	94139402	0.90	2	< 5	6	7.95	0.20	7.87	0.01	13.98	2.10	2.65	0.73	0.81	0.35	53.58	0.05	0.34	7.38	98.00
A0201-15	94139402	1.42	2	< 5	6	6.29	0.11	6.37	0.01	9.28	1.62	1.71	0.47	1.48	0.38	65.23	0.04	0.37	5.08	98.44
A0201-47	94139402	0.96	14	< 10	4	6.63	0.09	1.03	0.01	31.68	2.06	0.77	0.09	0.42	0.28	48.33	< 0.01	0.61	7.30	99.30
A0202-05	94139402	0.70	2	< 5	2	17.18	0.06	2.46	0.01	12.13	1.53	2.05	0.07	5.66	0.15	54.07	0.04	1.08	2.80	99.29
A0202-06	94139402	0.86	< 2	< 5	2	17.86	0.15	0.95	0.02	7.86	3.86	2.15	0.07	0.63	0.14	61.74	0.02	0.94	3.41	99.80
A0202-07	94139402	1.14	< 2	< 5	< 2	3.71	0.01	0.99	0.02	2.10	0.32	0.72	0.07	1.22	0.04	88.94	< 0.01	0.27	0.85	99.26
A0202-08	94139402	0.94	< 2	< 5	< 2	13.36	0.08	0.40	0.03	6.68	2.91	2.46	0.06	0.79	0.06	68.76	< 0.01	0.64	2.80	99.03
A0202-09	94139402	1.22	< 2	< 5	< 2	18.07	0.08	0.50	0.05	7.60	4.27	3.59	0.06	0.83	0.12	59.78	0.03	0.84	3.51	99.33
A0202-10	94139402	1.72	< 2	< 5	2	25.90	0.13	0.33	0.03	8.29	6.81	2.82	0.06	0.82	0.10	49.00	0.01	0.98	4.56	99.84
A0202-11	94139402	0.88	692	< 5	2	7.81	0.03	0.46	0.03	5.30	1.07	1.37	0.06	0.39	0.04	78.88	< 0.01	0.37	2.23	98.04
A0203-16	94139402	1.58	2	< 5	8	10.48	0.09	2.49	0.01	3.75	0.46	1.10	0.10	5.46	0.51	72.00	0.02	0.33	2.68	99.48
A0203-17	94139402	0.70	< 2	< 5	2	14.58	0.12	5.85	0.04	9.06	1.54	4.70	0.26	6.49	0.20	48.54	0.06	1.09	6.94	99.47
A0203-18	94139402	1.14	< 2	< 5	8	12.73	0.05	2.84	0.01	7.48	0.51	1.46	0.13	6.78	0.29	61.77	0.04	0.63	4.41	99.13
A0203-19	94139402	0.70	< 2	< 5	6	13.13	0.08	2.52	0.03	6.53	0.55	2.15	0.11	6.94	0.29	61.96	0.04	0.48	4.22	99.03
A0203-20	94139402	0.70	< 2	< 5	4	9.56	0.18	1.75	0.01	3.98	0.43	1.06	0.09	5.08	0.15	73.95	0.03	0.35	2.93	99.55
A0203-21	94139402	0.22	< 2	< 5	< 2	20.36	0.41	1.66	0.02	7.17	3.99	2.23	0.07	4.98	0.06	52.39	0.05	0.90	4.74	99.03
A0203-22	94139402	0.28	< 2	< 5	< 2	18.83	0.27	1.53	0.01	6.13	3.04	1.77	0.08	5.62	0.04	57.84	0.03	0.75	3.76	99.70
A0203-24	94139402	0.56	< 2	< 5	2	8.51	0.20	2.50	0.02	3.13	2.58	0.75	0.11	0.20	0.14	77.28	0.01	0.39	2.28	98.10
A0203-25	94139402	0.34	< 2	< 5	2	5.94	0.02	4.55	0.02	2.87	0.53	1.19	0.14	2.75	0.07	77.37	0.01	0.22	2.55	98.23
A0203-26	94139402	1.00	30	< 5	2	17.47	0.16	2.20	0.01	7.34	1.90	2.34	0.12	6.41	0.13	55.99	0.02	0.79	4.91	99.79
A0203-27	94139402	1.50	< 2	< 5	4	10.81	0.14	3.65	0.01	5.70	1.19	1.84	0.21	4.25	0.25	66.08	0.01	0.40	4.30	98.84
A0203-28	94139402	2.02	< 2	< 5	6	12.27	0.21	4.66	0.02	5.74	2.91	2.01	0.18	2.26	0.30	61.60	0.02	0.52	5.45	98.15
A0203-29	94139402	2.08	< 2	< 5	2	12.89	0.15	6.37	0.02	5.78	2.72	2.87	0.22	3.55	0.21	57.43	0.03	0.65	5.17	98.06
A0203-30	94139402	2.28	< 2	< 5	2	13.74	0.16	5.48	0.01	5.42	2.75	2.23	0.13	3.78	0.12	60.51	0.03	0.53	4.46	99.35
A0203-31	94139402	1.44	< 2	< 5	2	17.07	0.12	3.07	0.01	7.56	1.61	2.41	0.12	7.04	0.18	54.14	0.03	1.26	4.99	99.61
A0203-32	94139402	1.02	< 2	< 5	2	14.84	0.09	4.33	0.01	6.94	1.16	3.05	0.17	6.95	0.17	55.74	0.04	0.76	4.82	99.07
A0203-33	94139402	1.70	2	< 5	2	18.16	0.15	1.18	0.01	9.34	1.72	2.37	0.09	7.45	0.13	53.76	0.03	1.25	4.07	99.71
A0203-34	94139402	0.16	< 2	< 5	< 2	2.65	0.01	0.35	0.03	3.13	0.44	0.22	0.05	0.77	0.02	90.63	< 0.01	0.07	1.33	99.70
A0203-35	94139402	0.34	2	< 5	< 2	18.52	0.18	1.12	0.01	9.94	2.98	2.40	0.09	6.09	0.15	51.91	0.02	1.19	5.03	99.63
A0203-36	94139402	0.60	< 2	< 5	< 2	17.41	0.12	1.80	0.01	9.71	1.70	2.78	0.11	7.45	0.13	51.98	0.02	1.32	4.67	99.21
A0203-37	94139402	0.88	< 2	< 5	< 2	16.58	0.09	1.42	0.01	8.33	1.53	2.28	0.08	7.49	0.14	56.10	0.03	1.36	4.04	99.48
A0203-38	94139402	1.14	< 2	< 5	< 2	17.17	0.10	0.88	0.01	8.32	4.54	2.23	0.10	2.82	0.12	58.20	0.01	1.09	3.10	98.69
A0203-39	94139402	1.36	< 2	< 5	2	17.95	0.15	1.44	0.01	8.90	2.33	2.32	0.11	6.77	0.13	52.60	0.03	1.27	4.18	98.19
A0203-40	94139402	2.16	< 2	< 5	< 2	18.06	0.08	0.85	0.01	8.59	5.25	2.17	0.10	1.55	0.12	57.97	0.01	1.25	3.20	99.21
A0203-41	94139402	0.88	< 2	< 5	< 2	16.31	0.15	1.36	0.02	7.64	2.72	2.07	0.09	5.11	0.12	58.04	0.02	0.99	3.75	98.39
A0203-42	94139402	1.14	< 2	< 5	< 2	18.04	0.13	1.38	0.01	7.40	2.72	2.15	0.11	5.67	0.11	57.00	0.02	0.95	3.39	99.08

CERTIFICATION: [illegible signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Ag ppm	Ba ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm
A0200-51	413 402	< 1	1115	83.0	13.5	130	4.6	25	5.7	3.6	1.5	19	6.9	8	1.2	38.5	0.6	4	18	37.0
A0201-04	413 402	< 1	1610	105.0	20.0	170	12.3	270	7.1	4.2	1.8	24	8.9	5	1.5	50.5	0.5	4	24	49.0
A0201-12	413 402	< 1	820	110.5	20.0	140	4.6	25	4.6	2.8	1.4	23	7.1	5	1.0	53.0	0.4	6	19	44.5
A0201-13	413 402	< 1	160.0	36.0	19.0	290	1.4	225	1.5	1.0	0.4	6	2.3	4	0.3	18.0	0.1	6	9	13.5
A0201-14	413 402	< 1	1795	41.0	12.0	100	2.5	45	4.8	3.3	1.0	12	5.0	1	1.1	26.5	0.5	12	5	22.5
A0201-15	413 402	< 1	929	444	7.5	130	2.8	110	6.2	3.3	3.0	11	21.2	1	1.1	223	0.4	18	10	177.5
A0201-47	413 402	< 1	306	37.5	13.0	120	0.8	170	2.1	1.3	0.5	5	2.9	3	0.5	17.0	0.1	8	6	16.0
A0202-05	413 402	< 1	446	81.0	38.0	120	10.3	100	4.8	2.9	1.4	19	6.8	4	1.0	40.5	0.4	12	21	37.5
A0202-06	413 402	< 1	1340	95.5	21.0	160	3.4	50	5.4	3.2	1.4	24	7.4	4	1.2	47.5	0.4	6	19	42.5
A0202-07	413 402	< 1	62.0	28.0	9.5	160	0.4	20	1.5	0.9	0.4	4	2.1	4	0.3	14.5	0.1	6	4	11.0
A0202-08	413 402	< 1	552	96.0	16.5	170	3.6	45	4.5	2.7	1.4	19	6.8	4	1.0	47.5	0.4	6	11	40.5
A0202-09	413 402	< 1	738	106.5	16.5	190	5.7	35	5.6	3.6	1.6	24	7.2	6	1.2	51.0	0.5	6	14	42.5
A0202-10	413 402	< 1	1245	136.0	25.5	200	7.0	55	7.1	4.4	2.0	34	9.4	5	1.6	67.0	0.6	8	17	57.5
A0202-11	413 402	< 1	191.0	50.0	13.5	270	1.9	65	2.6	1.5	0.8	11	3.6	3	0.5	25.0	0.1	6	8	20.5
A0203-16	413 402	< 1	857	41.0	11.5	160	1.5	35	3.2	2.0	0.8	13	3.9	1	0.8	24.0	0.2	116	7	19.5
A0203-17	413 402	< 1	1055	40.5	28.0	320	7.3	65	2.8	2.2	0.8	22	3.6	1	0.7	22.0	0.4	26	15	19.0
A0203-18	413 402	< 1	380	53.5	16.0	140	1.7	70	3.2	2.0	1.0	14	4.3	1	0.7	30.5	0.3	102	13	25.0
A0203-19	413 402	< 1	769	68.5	12.0	160	1.3	70	4.2	2.5	1.1	15	5.5	1	0.9	39.5	0.3	104	10	31.0
A0203-20	413 402	< 1	1595	42.5	9.0	130	0.9	40	2.6	1.6	0.7	11	3.3	1	0.6	24.0	0.1	220	8	18.5
A0203-21	413 402	< 1	4410	123.0	17.5	160	5.0	65	4.3	2.5	1.6	31	7.3	3	0.9	63.0	0.3	8	16	53.0
A0203-22	413 402	< 1	2700	103.0	15.0	160	3.8	80	4.4	2.8	1.4	25	7.0	2	0.9	52.5	0.4	6	16	43.5
A0203-24	413 402	< 1	1820	47.5	7.0	180	3.3	55	1.9	1.2	0.6	15	3.2	1	0.4	27.0	0.1	102	9	19.0
A0203-25	413 402	< 1	289	15.0	4.5	190	1.7	25	2.1	1.5	0.4	9	1.9	1	0.5	10.0	0.2	78	3	7.5
A0203-26	413 402	< 1	1470	114.5	16.0	160	4.1	95	4.6	2.9	1.5	20	7.4	3	0.9	59.0	0.4	10	19	47.5
A0203-27	413 402	< 1	1315	63.0	11.5	160	3.1	80	4.2	2.7	0.9	15	5.1	3	0.9	32.5	0.4	22	11	27.5
A0203-28	413 402	< 1	2020	68.0	13.5	210	4.5	70	3.1	2.0	0.9	19	4.6	2	0.6	38.0	0.3	28	21	30.0
A0203-29	413 402	< 1	1450	60.5	12.0	190	13.0	60	3.4	2.1	1.0	17	4.4	3	0.7	34.0	0.3	16	17	27.0
A0203-30	413 402	< 1	1600	65.5	13.5	170	7.7	65	3.5	2.2	1.0	19	4.7	2	0.7	35.0	0.3	24	12	27.5
A0203-31	413 402	< 1	1080	93.5	18.5	160	2.7	75	3.9	2.3	1.4	21	6.6	5	0.8	46.5	0.4	8	25	41.0
A0203-32	413 402	< 1	700	97.5	14.5	170	3.9	90	3.8	2.3	1.4	19	6.4	3	0.7	51.5	0.4	14	16	42.0
A0203-33	413 402	< 1	1400	103.5	25.5	140	5.7	70	4.6	2.8	1.7	24	7.4	5	0.9	51.0	0.4	4	20	45.0
A0203-34	413 402	< 1	241	9.5	8.0	310	0.6	50	0.4	0.1	0.1	4	0.7	< 1	< 0.1	7.5	< 0.1	6	1	4.0
A0203-35	413 402	< 1	1755	102.5	27.0	160	6.4	85	5.1	2.8	1.8	25	7.4	4	1.0	50.5	0.4	6	23	45.0
A0203-36	413 402	< 1	1120	111.5	28.0	160	5.5	45	5.8	3.6	1.8	24	8.3	5	1.2	54.0	0.5	6	28	47.5
A0203-37	413 402	< 1	820	110.5	22.5	180	6.3	45	5.4	3.2	1.8	22	8.1	6	1.1	54.5	0.4	6	29	48.0
A0203-38	413 402	< 1	941	108.0	22.5	180	17.2	45	6.2	3.7	1.7	23	7.7	7	1.2	49.5	0.5	6	25	44.5
A0203-39	413 402	< 1	1540	109.5	25.0	160	10.9	80	5.3	3.1	1.8	25	7.9	5	1.0	54.5	0.4	6	23	46.5
A0203-40	413 402	< 1	800	107.5	23.0	200	15.7	55	6.2	3.7	1.7	25	8.0	6	1.2	53.5	0.5	6	27	46.5
A0203-41	413 402	< 1	1425	100.5	22.5	220	9.2	50	4.9	2.9	1.5	27	7.1	6	0.9	49.5	0.4	6	22	42.5
A0203-42	413 402	< 1	1195	119.0	20.5	170	8.7	50	5.3	3.1	1.7	25	7.7	6	1.1	56.5	0.4	6	30	46.0

CERTIFICATION: [signature]

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CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Ni ppm	Pb ppm	Pr ppm	Rb ppm	Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
A0200-51	94139402	25	10	9.8	127.0	7.1	3	132.0	1.5	1.1	11	< 0.5	0.6	2.5	70	3	35.0	3.6	70	322
A0201-04	94139402	50	60	13.0	184.5	9.3	5	212	1.5	1.3	12	0.5	0.6	3.0	110	23	42.0	3.4	100	176.0
A0201-12	94139402	35	10	12.8	149.0	8.0	3	220	1.5	1.0	14	0.5	0.4	2.0	75	3	26.0	2.8	95	165.5
A0201-13	94139402	30	270	4.0	52.8	2.6	5	26.2	0.5	0.3	5	< 0.5	0.1	0.5	25	77	9.0	0.9	15	142.5
A0201-14	94139402	35	5	5.9	74.0	4.6	1	416	0.5	0.8	4	< 0.5	0.5	3.5	150	4	33.0	3.0	125	60.0
A0201-15	94139402	45	< 5	51.0	61.6	24.9	1	379	0.5	2.0	15	< 0.5	0.4	4.0	130	3	33.5	2.4	120	74.0
A0201-47	94139402	35	580	4.3	34.2	3.2	1	38.4	< 0.5	0.4	4	< 0.5	0.1	0.5	20	9	12.5	1.2	25	98.0
A0202-05	94139402	55	25	10.5	117.5	7.4	3	411	1.5	1.0	9	< 0.5	0.4	3.0	155	7	27.0	2.7	90	164.5
A0202-06	94139402	35	5	11.6	158.0	8.1	3	169.5	1.5	1.1	11	< 0.5	0.5	2.5	90	10	31.0	3.0	85	144.0
A0202-07	94139402	25	5	3.1	15.0	2.2	< 1	39.4	< 0.5	0.3	5	< 0.5	0.1	1.0	15	6	8.5	0.9	20	157.5
A0202-08	94139402	50	20	11.1	124.5	7.2	3	87.6	0.5	0.9	14	< 0.5	0.4	3.5	70	13	25.0	2.4	70	167.5
A0202-09	94139402	45	20	12.3	176.5	7.9	4	104.5	1.0	1.1	15	< 0.5	0.6	3.5	95	10	33.0	3.4	90	202
A0202-10	94139402	65	25	16.3	256	10.5	6	122.5	1.5	1.4	20	0.5	0.7	4.5	145	17	41.5	4.0	95	165.0
A0202-11	94139402	35	20	5.8	54.0	4.0	3	70.4	0.5	0.5	7	< 0.5	0.2	1.5	30	18	13.5	1.2	40	108.5
A0203-16	94139402	60	75	5.4	29.0	3.8	3	298	< 0.5	0.6	6	< 0.5	0.3	8.5	115	8	23.0	1.7	695	50.0
A0203-17	94139402	105	45	5.3	107.0	3.7	4	478	1.0	0.5	5	< 0.5	0.4	4.0	400	12	23.5	2.3	535	77.0
A0203-18	94139402	65	50	7.1	29.2	4.5	2	406	0.5	0.6	6	< 0.5	0.3	7.5	170	7	21.5	1.8	425	68.5
A0203-19	94139402	65	60	8.8	30.0	5.6	1	404	0.5	0.8	8	< 0.5	0.4	8.5	145	6	27.0	2.1	380	63.0
A0203-20	94139402	55	40	5.2	22.4	3.3	2	286	0.5	0.5	6	< 0.5	0.2	9.0	95	5	16.5	1.4	110	53.5
A0203-21	94139402	50	20	15.0	183.0	8.8	7	304	1.5	1.0	16	< 0.5	0.3	3.5	505	12	24.0	2.1	160	88.5
A0203-22	94139402	50	15	12.4	131.5	7.3	7	240	1.0	0.9	13	< 0.5	0.4	3.5	300	13	26.0	2.4	140	71.5
A0203-24	94139402	35	10	5.7	143.5	3.3	15	94.3	1.0	0.4	7	< 0.5	0.1	3.5	185	11	11.5	1.1	25	50.5
A0203-25	94139402	40	35	2.0	34.8	1.5	2	223	< 0.5	0.3	4	< 0.5	0.2	2.5	90	3	16.0	1.3	145	39.5
A0203-26	94139402	45	10	14.0	113.0	8.4	7	235	1.5	0.9	18	< 0.5	0.4	4.0	215	11	26.5	2.3	95	115.5
A0203-27	94139402	40	10	7.9	77.8	5.1	4	222	0.5	0.7	10	< 0.5	0.4	4.5	220	6	26.0	2.5	115	115.5
A0203-28	94139402	60	10	8.5	165.0	5.1	17	225	30.0	0.6	10	< 0.5	0.3	6.5	315	9	19.0	1.7	155	79.0
A0203-29	94139402	45	10	7.6	177.5	4.7	10	323	3.5	0.6	9	0.5	0.3	2.5	190	7	21.5	1.9	115	101.0
A0203-30	94139402	50	5	7.9	136.0	4.7	7	330	3.5	0.6	10	< 0.5	0.3	4.5	265	8	21.0	2.0	125	76.5
A0203-31	94139402	40	10	11.4	65.8	7.3	4	322	2.0	0.8	12	< 0.5	0.3	2.5	255	10	21.0	2.2	60	179.5
A0203-32	94139402	40	15	12.0	62.2	7.3	3	384	1.5	0.9	14	< 0.5	0.3	3.5	260	8	20.5	2.1	150	125.5
A0203-33	94139402	45	15	12.5	93.4	8.2	3	273	1.5	1.0	13	< 0.5	0.4	2.0	225	7	23.5	2.3	155	167.5
A0203-34	94139402	15	5	1.2	19.2	0.8	1	39.7	0.5	< 0.1	1	< 0.5	< 0.1	< 0.5	10	4	2.0	0.1	15	17.0
A0203-35	94139402	45	10	12.4	141.0	8.5	6	215	3.5	1.0	14	< 0.5	0.4	2.0	265	10	25.0	2.4	140	158.0
A0203-36	94139402	50	10	13.1	88.4	8.7	4	276	2.0	1.1	14	< 0.5	0.5	2.5	265	8	32.0	3.0	120	165.5
A0203-37	94139402	45	10	13.4	81.0	8.8	4	282	2.0	1.1	15	< 0.5	0.5	2.5	225	7	29.0	2.8	125	217
A0203-38	94139402	45	5	12.2	209	8.5	3	136.5	2.0	1.1	15	0.5	0.5	2.0	110	3	33.0	3.2	120	235
A0203-39	94139402	50	15	13.3	128.0	9.0	3	302	1.5	1.1	15	< 0.5	0.4	2.0	225	4	27.5	2.6	145	202
A0203-40	94139402	45	5	13.0	225	8.6	3	132.5	2.0	1.2	16	0.5	0.5	2.0	110	2	33.0	3.3	120	227
A0203-41	94139402	50	5	12.0	140.0	7.9	4	243	1.5	1.0	14	< 0.5	0.4	2.0	165	7	25.5	2.5	120	214
A0203-42	94139402	45	5	13.1	133.0	8.3	3	232	2.0	1.0	15	< 0.5	0.4	2.0	105	4	28.0	2.9	170	222

CERTIFICATION: ____________

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SAMPLE	PREP CODE		Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)	Ge ppm (ICP)	Hf ppm	In ppm	K % (ICP)
A0200-51	9413	9402	0.06	7.53	< 0.2	1263.0	2.35	0.10	0.78	< 0.02	81.8	15.1	80	4.65	24.6	3.66	20.05	0.15	0.8	0.060	2.70
A0201-04	9413	9402	1.30	9.00	< 0.2	118.0	3.90	26.9	0.77	0.06	100.0	23.0	118	12.80	339.8	6.24	24.45	0.25	0.6	0.075	2.92
A0201-12	9413	9402	0.08	9.70	7.4	930.0	2.80	1.58	0.79	0.06	104.0	22.6	77	4.50	22.8	4.52	25.55	0.15	0.3	0.075	2.73
A0201-13	9413	9402	0.38	1.61	< 0.2	108.5	0.85	626	0.62	0.16	31.0	19.0	210	1.35	285.2	8.52	5.00	0.15	0.1	0.025	0.78
A0201-14	9413	9402	0.40	4.11	< 0.2	98.0	1.90	5.45	5.00	1.36	45.5	15.4	59	2.65	202.0	8.74	11.80	0.30	0.5	0.045	1.66
A0201-15	9413	9402	0.20	3.32	< 0.2	271.5	1.20	1.80	4.10	0.96	41.1	8.4	96	2.85	328.0	5.74	9.15	0.25	0.6	0.030	1.28
A0201-47	9413	9402	0.88	3.31	1.8	48.0	1.05	1.72	0.65	0.52	81.5	24.8	174	1.65	525.0	21.3	8.65	0.50	0.3	0.045	1.54
A0202-05	9413	9402	0.38	8.38	< 0.2	391.0	4.25	0.93	1.60	0.08	78.7	37.7	74	9.70	121.0	7.42	18.80	0.25	0.5	0.035	1.14
A0202-06	9413	9402	0.20	7.86	< 0.2	1129.0	2.50	0.56	0.50	0.02	87.0	20.4	100	3.10	68.2	4.40	21.10	0.25	0.2	0.065	2.64
A0202-07	9413	9402	0.04	1.73	< 0.2	56.0	0.35	0.25	0.51	0.04	26.4	7.0	99	0.40	13.2	1.19	3.25	0.15	< 0.1	0.005	0.24
A0202-08	9413	9402	0.14	7.05	< 0.2	528.9	2.35	0.57	0.25	0.02	97.0	17.9	96	3.65	45.6	4.28	19.50	0.25	0.4	0.055	2.30
A0202-09	9413	9402	0.10	9.71	< 0.2	731.6	2.60	0.33	0.34	< 0.02	111.0	19.0	105	5.95	42.0	4.89	25.50	0.25	0.4	0.075	3.40
A0202-10	9413	9402	0.16	12.15	< 0.2	1171.0	3.75	0.58	0.22	< 0.02	103.5	27.1	87	6.95	65.1	5.30	36.00	0.45	0.7	0.110	4.99
A0202-11	9413	9402	0.40	2.79	< 0.2	188.0	1.45	72.2	0.24	0.02	47.6	13.4	135	1.90	72.6	3.14	7.85	0.20	0.1	0.020	0.92
A0203-16	9413	9402	0.76	5.50	< 0.2	775.2	2.80	4.74	1.70	16.65	42.7	11.8	54	1.45	27.6	2.44	13.10	0.10	0.4	0.120	0.38
A0203-17	9413	9402	0.54	7.27	0.8	189.0	2.65	4.88	3.90	3.00	36.2	28.5	169	7.35	62.8	5.62	21.05	0.15	0.7	0.030	1.19
A0203-18	9413	9402	0.34	6.53	0.2	249.5	1.70	1.55	1.95	6.36	51.9	17.7	137	1.70	78.5	4.77	14.60	0.15	0.7	0.045	0.40
A0203-19	9413	9402	0.70	6.54	< 0.2	279.0	1.85	10.35	1.65	5.60	59.9	14.5	257	1.25	72.4	3.97	14.10	0.20	0.6	0.045	0.42
A0203-20	9413	9402	0.28	4.71	< 0.2	316.0	1.40	2.86	1.10	1.02	35.0	9.6	115	0.80	37.0	2.43	10.45	0.15	0.4	0.010	0.31
A0203-21	9413	9402	0.42	11.20	< 0.2	357.0	6.00	8.19	1.25	1.22	114.0	18.2	85	5.00	66.6	5.02	30.65	0.25	0.7	0.065	3.30
A0203-22	9413	9402	0.24	10.05	< 0.2	232.5	4.70	4.86	1.10	1.12	105.0	16.2	57	3.90	72.3	4.09	26.45	0.25	0.6	0.055	2.40
A0203-24	9413	9402	0.08	5.18	< 0.2	1580.0	6.15	4.77	1.95	0.32	44.9	7.1	72	3.20	42.8	2.37	14.95	0.20	0.4	0.045	2.37
A0203-25	9413	9402	0.26	3.15	< 0.2	261.5	1.20	1.44	3.10	1.42	13.40	4.0	72	1.65	14.8	1.89	8.30	0.10	0.3	0.020	0.43
A0203-26	9413	9402	0.20	9.36	< 0.2	365.0	4.05	5.64	1.60	0.64	106.0	17.2	86	4.35	89.4	4.97	20.85	0.20	0.8	0.035	1.55
A0203-27	9413	9402	0.18	5.58	< 0.2	423.5	3.05	1.64	2.40	1.06	66.6	11.3	62	3.10	94.9	3.64	14.90	0.20	1.2	0.030	0.96
A0203-28	9413	9402	0.14	6.43	< 0.2	330.5	5.90	2.40	3.20	1.70	61.8	17.1	312	4.65	77.1	3.81	18.90	0.20	0.7	0.050	2.30
A0203-29	9413	9402	0.14	6.68	< 0.2	339.5	4.70	1.28	4.40	0.84	63.4	13.8	191	13.00	54.1	3.70	17.55	0.20	1.1	0.035	2.19
A0203-30	9413	9402	0.18	7.02	< 0.2	703.7	4.10	1.12	3.70	0.98	62.2	14.1	95	7.50	67.1	3.45	19.10	0.15	0.8	0.040	2.15
A0203-31	9413	9402	0.26	8.21	< 0.2	331.5	2.50	2.68	2.00	0.28	100.5	20.5	106	2.95	87.6	4.64	22.70	0.20	0.9	0.045	1.25
A0203-32	9413	9402	0.30	8.13	< 0.2	641.4	2.20	1.83	3.20	0.88	91.9	15.4	64	3.90	81.4	4.66	19.55	0.20	1.1	0.040	0.98
A0203-33	9413	9402	0.24	9.08	< 0.2	114.5	2.65	1.44	0.82	0.30	96.0	25.4	62	5.65	70.1	5.97	23.35	0.20	0.7	0.065	1.33
A0203-34	9413	9402	0.06	1.37	< 0.2	216.0	0.55	0.51	0.24	0.02	7.81	8.6	126	0.55	28.4	2.10	3.00	0.05	< 0.1	0.005	0.34
A0203-35	9413	9402	0.20	10.35	0.4	96.0	3.60	1.57	0.87	0.28	106.5	30.5	106	7.40	92.7	7.16	27.90	0.30	0.9	0.090	2.55
A0203-36	9413	9402	0.28	8.94	< 0.2	135.5	2.60	1.83	1.25	0.28	107.0	28.9	94	5.60	41.0	6.45	24.95	0.25	0.9	0.060	1.37
A0203-37	9413	9402	0.16	8.43	0.2	136.5	2.15	1.87	1.00	0.20	107.5	24.9	114	6.40	31.4	5.44	22.80	0.25	0.6	0.055	1.21
A0203-38	9413	9402	0.06	9.01	< 0.2	873.6	2.25	0.23	0.62	0.12	99.6	26.1	233	16.75	51.7	5.63	24.05	0.30	0.6	0.070	3.58
A0203-39	9413	9402	0.16	9.00	< 0.2	132.0	3.05	0.76	0.98	0.20	96.8	27.3	163	10.85	68.3	5.76	25.75	0.35	0.6	0.080	1.84
A0203-40	9413	9402	0.06	10.25	< 0.2	761.2	2.35	0.14	0.63	0.06	108.0	25.8	150	15.75	49.0	6.33	25.95	0.45	0.5	0.075	4.45
A0203-41	9413	9402	0.10	9.65	< 0.2	221.5	3.20	0.57	1.05	0.14	98.2	28.7	285	9.15	39.6	5.84	26.65	0.25	0.4	0.085	2.45
A0203-42	9413	9402	0.12	9.58	< 0.2	598.6	2.65	0.28	0.96	0.12	113.5	23.1	90	9.10	40.8	5.11	27.20	0.30	0.6	0.075	2.24

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm	Sn ppm	Sr ppm (ICP)	Ta ppm (ICP)	Te ppm (ICP)
A0200-51	9413 9402	41.0	25.2	0.89	390	0.65	1.44	15.7	27.6	870	9.0	118.0	< 0.002	0.10	0.05	< 1	2.2	118.0	0.70	0.05
A0201-04	9413 9402	49.5	28.8	1.05	400	0.45	1.67	24.2	58.2	650	70.0	173.5	< 0.002	1.21	0.25	3	4.4	181.0	1.55	0.55
A0201-12	9413 9402	56.0	37.4	1.21	520	1.10	0.98	9.7	40.0	540	17.5	139.0	< 0.002	0.18	0.85	< 1	2.4	192.0	0.55	0.05
A0201-13	9413 9402	16.0	9.2	0.35	405	0.70	0.04	0.8	32.0	200	342	44.3	< 0.002	2.77	0.25	1	3.8	19.4	< 0.05	0.95
A0201-14	9413 9402	27.5	16.0	1.23	5020	5.20	0.57	1.5	68.9	1540	17.5	66.6	0.008	2.50	0.15	14	1.0	368	0.05	0.35
A0201-15	9413 9402	25.5	12.8	0.94	3180	9.85	0.56	1.7	62.7	1450	10.0	57.1	0.014	1.69	0.05	8	0.8	360	0.05	0.25
A0201-47	9413 9402	34.0	10.4	0.33	410	1.25	0.09	1.1	84.4	1270	568	63.3	< 0.002	6.48	0.10	3	1.0	69.4	< 0.05	0.15
A0202-05	9413 9402	40.0	20.8	0.92	415	5.85	4.29	6.7	56.8	620	28.0	100.0	0.006	1.92	0.05	5	1.2	356	0.45	0.20
A0202-06	9413 9402	44.5	20.8	0.87	310	1.25	0.35	12.5	38.4	550	10.5	126.0	0.002	0.53	0.05	2	2.6	127.0	0.50	0.10
A0202-07	9413 9402	12.5	4.6	0.28	280	0.85	0.80	2.3	21.2	220	6.5	12.5	< 0.002	0.13	< 0.05	< 1	0.2	31.4	< 0.05	< 0.05
A0202-08	9413 9402	49.5	28.4	1.16	255	2.00	0.54	7.1	60.6	330	18.5	112.0	0.002	0.42	< 0.05	1	2.4	76.3	0.20	< 0.05
A0202-09	9413 9402	54.0	33.4	1.51	325	1.55	0.57	6.5	56.1	520	22.5	163.0	0.002	0.32	< 0.05	1	3.2	99.8	0.20	< 0.05
A0202-10	9413 9402	47.0	38.6	1.40	285	0.85	0.53	8.3	76.4	350	29.0	215	0.002	0.43	< 0.05	1	4.0	108.0	0.40	< 0.05
A0202-11	9413 9402	24.0	7.6	0.41	235	0.80	0.13	4.7	35.4	130	30.5	50.4	< 0.002	0.96	< 0.05	< 1	1.8	41.6	< 0.05	0.25
A0203-16	9413 9402	24.0	19.6	0.57	510	92.38	4.15	0.9	69.7	2590	92.0	26.4	0.090	0.77	0.05	3	1.4	270	< 0.05	0.05
A0203-17	9413 9402	19.0	62.0	2.45	2040	19.85	4.94	1.6	123.5	840	47.5	95.2	0.020	2.61	0.25	5	2.0	454	0.10	0.05
A0203-18	9413 9402	30.0	16.4	0.75	870	89.76	5.21	2.3	112.5	1320	45.5	26.5	0.056	2.71	0.05	5	0.6	380	0.10	0.10
A0203-19	9413 9402	35.5	13.6	0.69	745	103.25	5.11	1.6	164.5	1290	73.5	26.2	0.068	1.91	0.10	5	0.8	351	0.10	0.10
A0203-20	9413 9402	20.5	9.8	0.49	495	171.75	3.58	1.1	84.8	630	33.5	18.0	0.102	1.27	0.10	3	0.8	244	0.05	0.05
A0203-21	9413 9402	55.5	50.4	1.29	550	3.25	3.93	2.4	55.0	220	26.5	163.5	0.004	1.46	0.15	6	5.6	268	0.15	0.05
A0203-22	9413 9402	52.0	35.4	1.01	545	1.60	4.45	2.3	54.9	170	15.0	132.5	0.002	1.06	0.05	5	5.6	226	0.15	0.05
A0203-24	9413 9402	25.5	19.4	0.50	755	84.36	0.14	1.3	38.2	680	8.0	132.5	0.094	0.64	0.05	1	13.2	91.2	< 0.05	< 0.05
A0203-25	9413 9402	7.5	12.6	0.71	900	55.82	2.02	0.5	40.4	340	34.5	30.5	0.062	1.13	< 0.05	1	1.2	201	< 0.05	< 0.05
A0203-26	9413 9402	56.0	33.2	1.29	820	6.10	5.16	3.9	58.8	560	11.0	105.0	0.010	1.88	0.05	4	5.2	216	0.25	0.05
A0203-27	9413 9402	36.0	30.4	0.99	1325	13.50	3.19	3.1	42.6	1070	9.0	69.9	0.012	1.05	< 0.05	4	2.8	196.0	0.20	< 0.05
A0203-28	9413 9402	35.0	51.0	1.12	1110	49.20	1.67	2.8	203	1290	10.5	149.0	0.046	1.07	0.05	3	13.4	201	0.15	< 0.05
A0203-29	9413 9402	35.0	58.4	1.60	1510	24.30	2.68	4.1	114.5	980	10.0	162.0	0.020	1.25	0.05	2	9.2	292	0.25	< 0.05
A0203-30	9413 9402	34.0	46.4	1.24	905	19.70	2.95	2.8	67.0	570	15.5	119.5	0.014	0.89	< 0.05	3	6.2	297	0.15	< 0.05
A0203-31	9413 9402	50.0	28.8	1.22	780	8.05	5.34	3.3	68.4	760	11.5	63.4	0.010	1.69	< 0.05	4	2.8	295	0.25	< 0.05
A0203-32	9413 9402	49.0	22.6	1.70	1305	8.50	5.73	1.9	42.2	740	25.5	55.4	0.008	1.47	< 0.05	3	1.8	371	0.15	< 0.05
A0203-33	9413 9402	44.0	32.4	1.20	575	0.75	5.78	2.7	50.2	560	16.5	81.2	< 0.002	2.35	< 0.05	4	1.8	227	0.20	0.05
A0203-34	9413 9402	4.0	5.4	0.16	150	0.95	0.55	0.3	13.0	80	3.0	17.0	< 0.002	1.23	0.05	1	0.8	36.2	< 0.05	< 0.05
A0203-35	9413 9402	47.5	43.0	1.34	590	3.15	4.83	3.0	67.4	740	12.0	141.5	0.004	3.75	0.05	5	4.0	202	0.20	0.05
A0203-36	9413 9402	50.0	31.6	1.43	735	2.70	5.92	3.0	62.8	590	15.0	79.2	0.004	2.98	< 0.05	4	2.2	238	0.20	< 0.05
A0203-37	9413 9402	52.5	28.2	1.16	565	4.95	5.95	3.6	64.7	620	10.5	75.6	0.006	2.90	< 0.05	3	1.6	249	0.25	< 0.05
A0203-38	9413 9402	49.5	48.8	1.22	595	19.50	2.11	12.3	129.5	510	7.0	193.5	0.018	0.64	0.05	1	2.2	125.0	0.35	0.05
A0203-39	9413 9402	47.0	39.2	1.16	745	11.40	5.28	9.3	102.5	570	13.0	118.0	0.010	2.37	< 0.05	4	2.4	240	0.65	0.05
A0203-40	9413 9402	54.0	58.0	1.28	665	5.95	1.26	7.5	78.6	580	7.0	206	0.002	0.42	< 0.05	1	2.0	124.5	0.20	< 0.05
A0203-41	9413 9402	47.0	38.2	1.26	595	22.65	4.32	9.6	170.0	560	8.5	125.5	0.006	1.51	0.20	1	2.6	228	0.50	0.05
A0203-42	9413 9402	55.5	40.4	1.21	735	2.00	4.53	25.2	58.6	580	7.0	124.0	< 0.002	0.95	0.05	1	2.0	219	1.70	< 0.05

CERTIFICATION:

ALS Chemex

Aurora Laboratory Services Ltd.
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212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
A0200-51	9413 9402	14.8	0.50	0.44	1.9	83	0.9	18.4	80	21.5	< 10
A0201-04	9413 9402	14.2	0.63	1.02	2.4	121	20.5	33.2	106	15.5	10
A0201-12	9413 9402	19.2	0.13	0.86	2.0	81	0.8	17.9	102	7.5	< 10
A0201-13	9413 9402	6.0	0.03	0.24	0.5	28	75.2	5.2	14	3.5	< 10
A0201-14	9413 9402	5.4	0.09	0.46	3.1	169	1.4	17.9	156	18.5	< 10
A0201-15	9413 9402	6.0	0.09	0.44	4.3	160	0.9	15.2	152	24.0	< 10
A0201-47	9413 9402	9.6	0.07	0.32	1.4	41	1.0	16.4	46	7.0	< 10
A0202-05	9413 9402	11.6	0.37	0.74	2.2	148	0.8	15.6	94	14.0	< 10
A0202-06	9413 9402	14.0	0.46	0.64	2.1	91	5.1	18.9	72	6.0	< 10
A0202-07	9413 9402	6.4	0.11	0.08	0.8	16	0.4	5.6	14	2.0	< 10
A0202-08	9413 9402	18.4	0.28	0.60	2.9	84	1.2	16.2	78	10.5	< 10
A0202-09	9413 9402	20.4	0.44	0.84	3.2	113	0.5	20.2	98	11.0	< 10
A0202-10	9413 9402	18.8	0.44	1.20	3.8	150	0.5	15.6	96	20.0	< 10
A0202-11	9413 9402	8.6	0.14	0.30	1.4	31	11.9	9.2	18	5.0	< 10
A0203-16	9413 9402	7.8	0.03	0.18	9.3	111	0.9	14.5	844	13.5	< 10
A0203-17	9413 9402	5.2	0.09	0.72	4.3	398	1.1	11.2	628	26.0	< 10
A0203-18	9413 9402	8.6	0.06	0.22	8.9	159	1.0	13.3	502	25.0	< 10
A0203-19	9413 9402	10.0	0.04	0.18	9.2	143	1.9	12.6	446	21.5	< 10
A0203-20	9413 9402	6.4	0.03	0.12	8.9	95	1.2	8.3	116	14.5	< 10
A0203-21	9413 9402	17.8	0.17	0.88	3.6	548	5.6	6.7	178	22.5	< 10
A0203-22	9413 9402	16.2	0.12	0.68	3.8	322	6.5	6.5	150	20.0	< 10
A0203-24	9413 9402	8.0	0.09	0.68	4.8	217	7.7	6.4	20	13.5	< 10
A0203-25	9413 9402	4.2	0.04	0.20	3.3	90	1.0	8.7	166	8.5	< 10
A0203-26	9413 9402	19.8	0.13	0.54	6.2	232	4.1	8.4	110	25.0	< 10
A0203-27	9413 9402	12.4	0.08	0.40	4.9	229	2.0	11.9	126	40.0	< 10
A0203-28	9413 9402	10.4	0.12	0.72	8.1	338	6.4	12.0	170	26.0	< 10
A0203-29	9413 9402	10.6	0.17	1.08	3.0	204	4.2	12.5	126	39.5	< 10
A0203-30	9413 9402	11.0	0.15	0.74	5.4	277	3.4	11.0	136	27.5	< 10
A0203-31	9413 9402	14.8	0.11	0.38	2.4	252	3.0	9.0	54	31.5	< 10
A0203-32	9413 9402	16.2	0.09	0.38	4.0	257	1.6	9.4	168	38.5	< 10
A0203-33	9413 9402	12.8	0.12	0.50	2.1	212	1.4	7.8	166	24.0	< 10
A0203-34	9413 9402	1.2	0.01	0.10	0.2	18	1.4	1.1	4	1.5	< 10
A0203-35	9413 9402	13.6	0.17	0.76	2.0	295	5.3	10.0	166	30.0	< 10
A0203-36	9413 9402	12.8	0.12	0.48	1.9	265	1.7	9.9	122	32.0	< 10
A0203-37	9413 9402	15.0	0.15	0.44	2.3	215	1.4	9.6	130	21.5	< 10
A0203-38	9413 9402	15.8	0.56	1.12	1.5	116	0.8	7.7	132	21.0	< 10
A0203-39	9413 9402	13.2	0.40	0.76	1.6	217	1.4	8.7	146	20.0	< 10
A0203-40	9413 9402	16.2	0.66	1.16	1.5	115	0.1	8.6	132	15.5	< 10
A0203-41	9413 9402	14.6	0.28	0.78	2.2	192	2.6	7.5	146	13.0	< 10
A0203-42	9413 9402	17.2	0.45	0.74	1.9	119	1.7	10.4	192	20.5	< 10

CERTIFICATION: [signature]

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212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Weight Kg	Au ppb ICP	Pt ppb ICP	Pd ppb ICP	Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %
A0203-43	9413 9402	1.02	< 2	< 5	< 2	18.27	0.18	1.44	0.01	9.04	2.30	2.27	0.08	6.87	0.15	53.58	0.04	1.19	4.40	99.82
A0203-44	9413 9402	1.28	< 2	< 5	4	13.84	0.15	3.51	0.01	6.64	1.88	1.97	0.14	5.14	0.41	58.91	0.03	0.55	5.48	98.66
A0203-45	9413 9402	2.02	< 2	< 5	6	11.63	0.17	5.35	0.01	5.14	2.34	2.43	0.18	3.22	0.28	61.83	0.03	0.49	5.04	98.14
A0203-46	9413 9402	1.96	< 2	< 5	2	11.20	0.20	5.60	0.01	6.04	3.51	2.44	0.18	0.65	0.24	60.67	0.03	0.82	6.63	98.22
A0203-49	9413 9402	1.50	< 2	< 5	< 2	18.06	0.12	0.60	0.01	8.31	3.56	2.44	0.07	1.41	0.10	60.30	0.03	1.15	3.76	99.92
A0204-01	9413 9402	1.28	14	5	14	13.51	0.12	2.02	0.01	17.25	0.97	1.06	0.17	5.61	0.13	51.72	0.02	0.87	6.17	99.63
A0204-02	9413 9402	1.72	42	< 5	26	11.97	0.10	3.15	0.01	14.02	0.79	1.27	0.18	5.31	0.22	53.43	0.02	0.65	6.92	98.04
A0204-03	9413 9402	1.98	10	< 5	6	15.55	0.21	3.84	0.01	19.47	4.81	2.58	0.11	0.42	0.09	38.76	0.02	0.99	11.17	98.03
A0204-48	9413 9402	0.86	< 2	< 5	4	13.38	0.25	3.86	0.02	5.65	2.31	1.86	0.12	3.90	0.25	62.08	0.02	0.61	5.30	99.61
A0204-50	9413 9402	0.34	34	15	28	4.22	0.11	4.57	0.01	28.92	1.11	2.03	1.37	0.10	1.46	35.51	0.01	0.20	18.60	98.22
A0204-52	9413 9402	0.66	36	< 5	4	11.93	0.30	6.64	0.02	5.56	3.13	1.77	0.17	1.24	0.23	56.99	0.04	0.40	9.62	98.04
A0204-53	9413 9402	0.28	526	5	4	4.56	0.14	7.55	0.01	15.25	1.32	2.33	1.00	0.13	0.17	52.55	0.04	0.22	13.47	98.74
A0204-54	9413 9402	0.42	16	< 5	4	15.13	0.73	1.43	0.03	4.65	4.14	1.29	0.06	1.36	0.14	62.93	0.02	0.48	7.37	99.76
A0204-55	9413 9402	1.28	< 2	< 5	< 2	18.67	0.13	1.28	0.01	7.47	2.84	2.37	0.08	4.57	0.15	57.34	0.01	0.82	3.79	99.53
A9801-0201	9413 9402	1.62	< 2	< 5	2	13.62	0.21	4.07	0.02	5.00	2.49	1.70	0.07	1.70	0.14	63.69	0.05	0.56	5.92	99.24
A9801-0202	9413 9402	2.22	2	< 5	4	11.39	0.17	6.39	0.02	3.99	2.42	1.43	0.07	1.39	0.14	62.74	0.05	0.43	7.49	98.12
A9801-0203	9413 9402	2.04	< 2	< 5	2	15.55	0.19	4.32	0.01	5.94	3.43	1.94	0.09	2.48	0.19	57.82	0.05	0.77	6.23	99.01
A9801-0204	9413 9402	1.28	2	< 5	2	8.81	0.03	6.84	0.01	2.95	0.33	1.29	0.08	6.10	0.09	67.25	0.06	0.32	4.34	98.50
A9801-0205	9413 9402	1.44	4	< 10	< 4	7.25	0.03	4.84	0.02	2.74	0.67	0.85	0.08	3.86	0.06	74.63	0.04	0.31	3.12	98.50
A9801-0207	9413 9402	1.30	4	< 5	4	12.09	0.23	5.16	0.02	4.96	3.03	1.49	0.11	1.24	0.18	61.40	0.04	0.50	8.95	99.40
A9802-06	9413 9402	1.40	2	< 5	4	10.39	0.12	10.45	0.01	4.25	1.62	1.95	0.13	3.99	0.18	56.25	0.06	0.42	8.75	98.57
1LF05-12	9413 9402	0.68	114	< 10	< 4	3.33	0.01	0.34	0.01	19.64	0.20	1.97	0.26	0.27	0.21	63.91	< 0.01	0.15	9.18	99.48
1LF05-13	9413 9402	0.96	6	< 5	< 2	7.26	0.01	0.16	< 0.01	16.95	0.43	4.58	0.22	0.29	0.07	60.52	< 0.01	0.36	8.95	99.80
1LF05-14	9413 9402	0.08	100	< 5	6	5.42	< 0.01	0.04	0.01	29.89	0.14	2.46	0.10	0.32	0.05	49.13	< 0.01	0.32	10.21	98.09
1LF05-17	9413 9402	0.56	4	< 5	< 2	8.32	0.06	0.10	0.01	9.93	1.82	2.28	0.21	0.53	0.03	68.33	< 0.01	0.39	7.80	99.81
1LF05-18	9413 9402	0.58	10	< 5	< 2	9.76	0.08	0.18	0.01	13.63	2.07	2.35	0.21	0.68	0.03	61.40	< 0.01	0.53	9.04	99.97
1LF05-19	9413 9402	1.32	< 2	< 5	< 2	8.88	0.09	0.29	0.01	6.27	2.08	1.08	0.11	0.47	0.03	75.26	< 0.01	0.46	4.56	99.59
1LF05-20	9413 9402	0.42	< 2	< 5	4	3.61	< 0.01	17.36	0.19	8.65	0.06	15.19	0.36	0.06	0.08	25.55	0.04	0.53	27.66	99.34
1LF05-23	9413 9402	1.44	68	< 10	4	5.88	0.03	0.17	0.01	14.19	0.55	3.27	0.18	0.55	0.06	62.87	< 0.01	0.32	10.42	98.50
1LF05-24	9413 9402	1.02	54	< 5	2	6.87	0.03	0.44	0.01	13.24	0.79	3.52	0.21	0.69	0.13	61.48	< 0.01	0.34	10.87	98.62
1LF05-25	9413 9402	1.84	38	< 5	< 2	7.70	0.05	0.19	0.01	11.08	1.04	3.04	0.17	0.61	0.06	65.99	< 0.01	0.36	8.00	98.30
1LF05-26	9413 9402	2.22	30	< 5	2	7.31	0.03	0.19	0.01	11.06	0.90	3.06	0.17	0.75	0.09	66.31	< 0.01	0.37	8.77	99.02
1LF05-27	9413 9402	1.24	6	< 5	< 2	12.48	0.11	0.15	0.01	9.08	2.74	2.75	0.17	0.99	0.06	61.65	< 0.01	0.59	8.42	99.20
1LF05-30	9413 9402	1.02	6	< 5	< 2	6.12	0.01	0.10	0.01	12.65	0.63	4.10	0.22	0.47	0.02	66.69	< 0.01	0.20	8.74	99.96
1LF06-01	9413 9402	0.76	14	< 5	2	5.90	0.03	0.50	< 0.01	25.92	0.97	3.59	0.37	0.24	0.08	46.13	< 0.01	0.37	15.20	99.30
1LF06-07	9413 9402	0.50	132	< 5	2	11.86	0.07	0.58	< 0.01	15.17	2.54	4.08	0.24	0.85	0.05	50.22	< 0.01	0.37	12.90	98.93
1LF06-08	9413 9402	0.48	36	< 5	< 2	16.64	0.10	0.19	< 0.01	25.15	3.60	6.31	0.39	1.18	0.06	24.38	0.02	0.73	21.09	99.84
1LF06-10	9413 9402	0.54	< 2	< 5	< 2	11.71	0.05	0.84	0.01	3.24	2.66	1.28	0.07	0.58	0.08	74.53	< 0.01	0.49	4.19	99.73
1LF06-11	9413 9402	0.80	6	< 5	< 2	14.61	0.08	0.56	0.01	4.20	3.84	2.12	0.07	0.46	0.25	66.08	< 0.01	0.61	5.15	98.04
1LF06-21	9413 9402	0.22	216	< 5	< 2	7.98	0.05	0.22	< 0.01	34.41	1.55	8.43	0.32	0.79	0.05	18.11	< 0.01	0.36	27.22	99.49

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
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CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE		Ag ppm	Ba ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm
A0203-43	413	402	< 1	1650	104.5	26.5	150	5.8	85	5.3	3.1	1.7	26	7.8	5	1.0	50.5	0.4	6	24	45.5
A0203-44	413	402	< 1	1390	69.5	14.5	160	2.8	85	5.2	3.2	1.2	21	5.8	2	1.0	36.5	0.4	30	18	30.0
A0203-45	413	402	< 1	1570	70.0	14.0	190	6.1	75	3.7	2.2	1.2	17	5.2	2	0.7	40.5	0.3	44	12	31.0
A0203-46	413	402	< 1	1745	55.0	16.5	170	3.6	105	3.5	2.3	1.0	16	4.5	3	0.7	29.5	0.3	22	21	24.5
A0203-49	413	402	< 1	999	113.5	23.0	200	3.5	115	6.6	3.8	2.0	25	8.8	4	1.3	56.5	0.5	6	14	49.5
A0204-01	413	402	< 1	991	86.0	30.5	200	1.4	210	6.2	4.0	1.4	16	7.8	3	1.3	52.0	0.5	42	21	43.0
A0204-02	413	402	< 1	918	61.0	36.5	200	1.2	445	4.0	2.7	1.0	15	5.2	3	0.9	36.0	0.4	72	16	28.5
A0204-03	413	402	< 1	2000	92.5	129.5	210	5.8	185	4.4	2.3	1.6	23	6.9	4	0.8	46.0	0.4	14	16	40.5
A0204-48	413	402	< 1	2550	61.0	16.5	250	2.6	70	3.5	2.3	1.0	20	4.8	2	0.7	34.0	0.3	46	12	27.0
A0204-50	413	402	< 1	429	20.0	24.5	140	1.4	265	5.1	3.7	1.0	4	4.8	< 1	1.2	21.0	0.5	12	3	17.5
A0204-52	413	402	< 1	1415	28.0	6.5	120	2.7	20	2.0	1.3	0.5	9	2.4	1	0.4	17.0	0.1	14	5	12.5
A0204-53	413	402	< 1	1125	42.5	11.5	240	2.2	70	5.5	3.5	1.0	8	6.1	1	1.2	31.5	0.5	8	5	28.0
A0204-54	413	402	< 1	6720	67.5	14.0	300	7.2	95	3.1	1.9	1.2	21	4.9	1	0.6	39.0	0.3	28	9	30.0
A0204-55	413	402	< 1	1260	117.5	19.0	190	4.0	125	5.6	3.2	1.7	25	8.4	4	1.1	58.0	0.4	24	14	50.0
A9801-0201	413	402	< 1	1900	78.0	14.5	210	3.0	25	3.9	2.5	1.2	19	5.5	4	0.8	40.5	0.4	18	12	34.0
A9801-0202	413	402	< 1	1505	55.5	10.5	200	4.6	40	3.0	1.9	1.0	16	4.0	1	0.6	31.0	0.3	20	8	24.0
A9801-0203	413	402	< 1	1810	99.5	17.0	240	5.6	65	5.4	3.3	1.5	21	7.4	4	1.1	48.5	0.4	20	17	42.5
A9801-0204	413	402	< 1	301	49.0	9.5	200	0.7	50	2.7	2.2	0.8	10	3.6	1	0.6	27.5	0.4	48	7	21.5
A9801-0205	413	402	< 1	283	37.0	9.0	250	1.6	35	2.0	1.1	0.6	9	2.9	1	0.4	21.5	0.1	62	7	16.5
A9801-0207	413	402	< 1	2050	65.0	13.0	230	5.4	25	3.6	2.3	1.2	17	4.9	1	0.8	35.5	0.3	28	11	28.5
A9802-06	413	402	< 1	1305	53.5	10.5	200	2.8	30	3.7	2.5	1.0	14	4.4	2	0.9	28.0	0.4	30	10	24.5
1LF05-12	413	402	4	27.0	9.5	37.0	140	0.1	>10000	2.0	1.3	0.1	3	1.7	< 1	0.4	6.0	0.1	8	1	5.0
1LF05-13	413	402	1	119.5	66.0	21.5	130	0.5	700	3.8	2.5	0.7	12	4.8	2	0.8	30.0	0.3	8	7	26.5
1LF05-14	413	402	5	19.0	27.0	28.5	120	0.1	5680	2.6	1.7	0.3	5	2.8	< 1	0.6	14.0	0.2	8	3	12.5
1LF05-17	413	402	< 1	521	70.0	13.0	190	1.5	290	2.9	1.7	0.8	11	4.4	6	0.6	31.0	0.2	8	7	27.0
1LF05-18	413	402	1	646	81.5	29.0	190	1.6	440	3.3	2.0	0.8	14	5.1	8	0.7	38.0	0.3	8	10	32.5
1LF05-19	413	402	< 1	750	91.5	8.0	220	1.7	110	3.4	2.0	0.7	11	5.5	9	0.7	39.5	0.3	8	9	34.5
1LF05-20	413	402	< 1	36.5	15.0	65.0	1600	0.6	45	1.2	0.7	0.6	5	1.6	< 1	0.3	8.0	< 0.1	6	7	7.0
1LF05-23	413	402	3	142.5	59.0	28.0	210	0.6	1755	3.9	2.5	0.4	10	4.7	1	0.9	26.0	0.3	10	6	24.0
1LF05-24	413	402	< 1	233	65.5	21.0	230	0.9	300	4.1	2.5	0.5	12	5.2	1	0.9	30.0	0.3	8	7	26.5
1LF05-25	413	402	2	336	69.5	16.5	230	1.1	465	3.6	2.3	0.4	12	4.9	1	0.8	31.0	0.3	8	8	28.0
1LF05-26	413	402	1	271	75.5	15.5	210	0.9	705	3.8	2.5	0.5	12	5.3	1	0.8	33.5	0.3	8	8	30.0
1LF05-27	413	402	< 1	773	78.5	10.5	190	2.3	280	3.5	2.3	0.8	18	5.1	6	0.8	36.0	0.3	8	12	32.5
1LF05-30	413	402	< 1	126.5	37.5	12.5	150	0.5	195	2.2	1.3	0.4	9	2.8	4	0.5	18.5	0.1	8	4	15.5
1LF06-01	413	402	< 1	238	67.5	51.0	190	1.1	1300	3.2	1.8	0.7	10	4.6	1	0.7	31.5	0.2	8	8	27.0
1LF06-07	413	402	< 1	680	78.5	39.0	160	1.9	1340	2.4	1.4	0.8	18	4.5	5	0.5	34.0	0.2	8	7	31.0
1LF06-08	413	402	< 1	832	167.5	27.5	120	2.5	795	4.5	2.7	1.5	23	9.1	12	0.9	74.0	0.4	6	11	63.0
1LF06-10	413	402	< 1	495	84.5	8.0	220	2.0	25	3.2	1.9	1.0	16	5.2	6	0.7	38.5	0.3	8	11	34.0
1LF06-11	413	402	< 1	701	102.5	11.5	190	3.3	20	5.6	3.2	1.5	21	8.0	7	1.2	48.0	0.4	8	14	43.5
1LF06-21	413	402	2	303	32.5	51.0	110	1.4	205	1.3	0.9	0.3	11	2.3	3	0.3	17.0	0.1	8	4	14.0

CERTIFICATION:

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CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Ni ppm	Pb ppm	Pr ppm	Rb ppm	Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
A0203-43	9413 9402	50	20	12.6	109.0	8.5	5	274	1.5	1.1	14	< 0.5	0.4	2.5	265	7	27.5	2.5	165	182.0
A0203-44	9413 9402	45	15	8.6	79.6	5.5	4	284	1.0	0.9	11	< 0.5	0.5	5.5	325	5	30.5	2.9	90	79.5
A0203-45	9413 9402	55	30	9.0	104.5	5.5	2	355	0.5	0.7	10	< 0.5	0.3	4.5	190	3	21.5	1.9	155	89.0
A0203-46	9413 9402	45	20	6.9	134.5	4.7	2	236	1.5	0.7	7	< 0.5	0.3	3.0	180	3	21.5	2.0	125	107.0
A0203-49	9413 9402	45	15	14.1	155.5	9.2	4	155.0	1.5	1.3	16	< 0.5	0.5	2.5	125	4	34.5	3.0	100	158.0
A0204-01	9413 9402	160	35	11.8	41.0	7.6	4	226	1.5	1.1	13	< 0.5	0.5	12.5	350	12	39.5	3.3	30	123.5
A0204-02	9413 9402	135	65	8.2	34.0	5.3	3	265	0.5	0.7	9	< 0.5	0.4	9.5	230	9	25.5	2.2	190	91.0
A0204-03	9413 9402	165	30	11.0	195.5	7.6	5	215	1.5	0.9	15	0.5	0.3	2.0	120	28	21.0	2.1	330	150.0
A0204-48	9413 9402	55	25	7.7	99.2	4.9	5	281	0.5	0.7	10	< 0.5	0.3	5.0	240	9	22.5	2.1	50	82.5
A0204-50	9413 9402	180	20	4.2	36.8	3.8	< 1	100.5	< 0.5	0.8	3	< 0.5	0.5	3.5	165	3	38.0	3.0	110	33.0
A0204-52	9413 9402	30	25	3.6	75.6	2.4	1	200	< 0.5	0.4	4	< 0.5	0.1	2.5	135	3	11.5	1.2	130	51.5
A0204-53	9413 9402	55	25	7.0	68.6	5.5	1	371	< 0.5	1.0	4	< 0.5	0.5	2.0	105	5	32.0	3.1	55	57.0
A0204-54	9413 9402	55	5	8.5	182.0	5.2	2	247	0.5	0.6	11	0.5	0.3	8.0	555	4	17.5	1.7	555	66.5
A0204-55	9413 9402	45	15	13.9	134.5	9.2	3	221	1.5	1.1	19	< 0.5	0.4	4.0	240	3	27.5	2.6	150	135.0
A9801-0201	9413 9402	45	15	9.6	109.0	6.0	3	476	0.5	0.8	12	0.5	0.3	5.0	315	5	22.0	2.2	155	138.5
A9801-0202	9413 9402	45	10	7.0	107.0	4.3	2	490	0.5	0.6	9	0.5	0.3	4.5	280	7	18.5	1.6	210	66.5
A9801-0203	9413 9402	50	60	11.6	161.0	7.8	4	524	1.0	1.0	15	0.5	0.5	4.0	210	6	32.0	2.8	120	169.5
A9801-0204	9413 9402	65	140	6.2	17.2	3.9	1	716	< 0.5	0.5	7	< 0.5	0.4	6.0	275	8	22.5	2.3	170	63.5
A9801-0205	9413 9402	55	200	4.5	35.0	2.9	3	445	< 0.5	0.4	6	< 0.5	0.1	5.5	115	8	13.0	0.9	40	55.5
A9801-0207	9413 9402	45	10	8.3	149.5	5.1	3	459	0.5	0.7	10	0.5	0.3	6.5	405	5	22.5	1.9	55	69.5
A9802-06	9413 9402	50	105	6.5	70.2	4.4	2	647	0.5	0.6	8	< 0.5	0.4	6.5	340	4	25.0	2.5	490	67.0
1LF05-12	9413 9402	25	480	1.1	3.2	1.3	69	13.2	< 0.5	0.3	1	< 0.5	0.1	2.0	35	7	13.5	1.0	455	22.0
1LF05-13	9413 9402	30	260	7.2	19.6	4.7	132	24.4	0.5	0.7	9	< 0.5	0.4	2.0	45	13	21.5	2.3	840	65.5
1LF05-14	9413 9402	35	1020	3.2	2.8	2.5	87	11.4	< 0.5	0.4	3	< 0.5	0.3	4.0	55	7	18.0	1.5	1965	29.0
1LF05-17	9413 9402	20	50	7.3	74.4	4.8	28	54.3	0.5	0.6	13	< 0.5	0.3	2.0	30	7	14.5	1.7	290	234
1LF05-18	9413 9402	20	180	8.8	82.4	5.6	26	48.6	0.5	0.7	17	< 0.5	0.3	2.0	40	8	17.0	1.9	210	277
1LF05-19	9413 9402	15	40	9.6	88.8	5.9	42	40.2	0.5	0.7	20	< 0.5	0.3	2.5	25	9	17.0	1.9	90	330
1LF05-20	9413 9402	685	10	1.8	2.4	1.7	2	394	< 0.5	0.2	< 1	< 0.5	< 0.1	1.0	45	6	6.5	0.5	75	49.5
1LF05-23	9413 9402	40	350	6.6	24.0	4.4	50	57.3	< 0.5	0.7	8	< 0.5	0.4	3.0	75	13	24.5	2.2	930	53.0
1LF05-24	9413 9402	40	255	7.2	34.2	5.0	30	85.3	0.5	0.7	8	< 0.5	0.4	2.0	40	11	23.5	2.1	1075	53.0
1LF05-25	9413 9402	30	495	7.5	46.6	4.8	42	51.1	0.5	0.7	9	< 0.5	0.3	1.5	40	11	20.5	2.1	1065	59.0
1LF05-26	9413 9402	25	145	8.1	39.8	5.4	35	68.9	0.5	0.7	10	< 0.5	0.3	1.5	45	11	21.0	2.2	460	62.0
1LF05-27	9413 9402	30	110	8.7	114.0	5.4	66	71.8	0.5	0.7	15	< 0.5	0.3	2.0	60	9	18.5	2.1	335	203
1LF05-30	9413 9402	20	115	4.2	26.8	2.7	14	41.6	< 0.5	0.4	7	< 0.5	0.1	0.5	15	4	12.0	1.1	145	130.5
1LF06-01	9413 9402	55	260	7.1	46.8	4.9	29	28.4	0.5	0.6	7	< 0.5	0.3	2.0	40	9	16.5	1.6	550	56.0
1LF06-07	9413 9402	20	70	8.2	106.0	4.9	44	95.2	0.5	0.5	12	< 0.5	0.2	2.0	40	8	12.0	1.3	320	159.0
1LF06-08	9413 9402	15	60	17.7	134.5	10.2	50	134.0	0.5	1.0	27	0.5	0.4	3.5	45	28	22.0	2.5	385	427
1LF06-10	9413 9402	25	30	9.2	115.5	5.7	3	72.8	0.5	0.6	17	< 0.5	0.3	2.5	45	5	16.5	1.9	40	216
1LF06-11	9413 9402	30	25	11.6	174.5	8.1	5	57.1	1.0	1.1	19	0.5	0.5	4.5	75	5	29.0	2.8	50	254
1LF06-21	9413 9402	45	1190	3.8	63.0	2.4	27	79.6	< 0.5	0.3	8	0.5	0.1	1.0	35	6	7.5	0.9	4390	110.0

CERTIFICATION: [signature]

ALS Chemex

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To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE		Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)	Ge ppm (ICP)	Hf ppm	In ppm	K % (ICP)
A0203-43	9413	9402	0.26	9.22	< 0.2	166.0	3.05	1.87	1.00	0.52	100.5	27.0	73	6.35	91.8	5.76	27.00	0.25	0.8	0.090	1.81
A0203-44	9413	9402	0.14	7.82	< 0.2	289.0	2.85	1.00	2.50	0.52	65.9	13.6	80	2.95	97.2	4.58	21.45	0.20	0.7	0.070	1.58
A0203-45	9413	9402	0.18	6.34	< 0.2	185.0	2.20	0.70	3.70	0.76	68.5	14.4	85	6.35	63.6	3.41	16.85	0.20	0.9	0.040	1.93
A0203-46	9413	9402	0.18	5.90	< 0.2	568.4	2.15	0.67	3.80	0.60	59.9	16.7	99	3.80	88.4	3.94	16.65	0.25	1.2	0.045	2.75
A0203-49	9413	9402	0.10	8.91	< 0.2	924.1	2.90	0.16	0.41	0.06	101.5	22.8	107	3.65	129.1	5.29	23.60	0.25	0.6	0.080	2.65
A0204-01	9413	9402	1.42	6.88	5.4	71.0	2.10	29.4	1.40	0.32	77.5	30.5	151	1.40	486.0	11.50	15.40	0.35	1.3	0.020	0.75
A0204-02	9413	9402	1.64	6.33	69.2	92.0	1.75	31.5	2.10	2.26	58.1	38.1	243	1.20	571.0	9.24	13.65	0.25	1.0	0.030	0.62
A0204-03	9413	9402	0.56	8.20	< 0.2	178.5	2.20	2.07	2.60	1.16	80.5	119.4	107	5.75	219.6	12.75	22.30	0.35	0.5	0.090	3.56
A0204-48	9413	9402	0.20	7.32	< 0.2	990.0	3.25	1.86	2.80	0.30	53.6	15.3	99	2.65	74.9	3.85	19.10	0.15	0.7	0.040	1.91
A0204-50	9413	9402	1.44	2.23	42.8	31.5	0.65	2.02	3.10	0.48	35.9	44.1	110	2.65	663.0	19.80	6.80	0.40	0.4	0.030	0.93
A0204-52	9413	9402	0.28	6.21	35.4	301.5	2.20	1.70	4.40	2.42	44.1	10.5	110	5.35	27.6	3.56	15.45	0.10	1.2	0.040	2.42
A0204-53	9413	9402	0.36	2.59	318	32.0	1.10	2.50	5.50	0.62	38.9	11.8	122	2.55	113.4	10.55	7.35	0.25	0.5	0.035	1.13
A0204-54	9413	9402	0.18	9.12	23.2	137.0	3.15	0.67	1.10	7.50	52.6	13.8	135	7.70	127.2	3.53	22.35	0.15	0.7	0.060	3.65
A0204-55	9413	9402	0.18	9.91	< 0.2	596.4	3.35	0.51	0.91	0.30	111.0	19.5	99	4.25	81.9	5.02	25.85	0.25	0.9	0.075	2.26
A9801-0201	9413	9402	0.18	7.50	< 0.2	923.2	2.80	0.13	2.90	0.60	73.6	15.1	94	3.10	49.8	3.35	19.30	0.20	0.5	0.045	2.08
A9801-0202	9413	9402	0.10	6.20	< 0.2	529.9	2.10	0.42	4.50	2.46	53.7	13.3	158	4.55	50.9	2.67	15.10	0.15	0.4	0.055	1.96
A9801-0203	9413	9402	0.40	7.95	< 0.2	415.5	3.00	1.87	2.90	0.68	91.9	20.9	207	5.40	71.2	3.74	20.55	0.25	0.6	0.060	2.57
A9801-0204	9413	9402	0.50	4.52	< 0.2	271.0	1.40	2.08	4.90	2.42	45.4	8.8	72	0.70	52.4	1.94	8.95	0.10	0.3	0.025	0.27
A9801-0205	9413	9402	1.14	3.66	< 0.2	249.0	1.35	4.60	3.40	0.56	35.6	9.3	97	1.55	33.8	1.82	7.95	0.05	0.3	0.010	0.52
A9801-0207	9413	9402	0.16	6.67	1.4	233.5	2.80	1.39	3.70	0.34	57.7	12.3	97	5.40	59.3	3.36	16.20	0.15	0.5	0.040	2.51
A9802-06	9413	9402	1.22	5.48	< 0.2	1174.0	2.45	4.71	7.70	6.52	51.2	10.9	83	2.80	69.8	2.74	14.40	0.15	0.5	0.080	1.32
1LF05-12	9413	9402	15.40	1.64	352	46.0	0.15	25.6	0.23	4.38	15.35	73.2	156	0.20	>10000	13.10	4.40	0.25	0.6	11.25	0.11
1LF05-13	9413	9402	2.10	4.08	70.8	105.0	0.25	4.49	0.11	3.18	62.3	22.3	114	0.50	829.0	11.40	11.85	0.20	1.2	1.485	0.33
1LF05-14	9413	9402	25.7	2.99	515	29.0	0.15	75.1	0.03	14.55	48.1	54.5	130	0.20	7130	20.6	7.65	0.40	1.1	19.00	0.09
1LF05-17	9413	9402	0.94	4.83	23.6	482.5	0.75	2.11	0.06	1.06	65.5	13.5	93	1.45	381.0	7.22	11.25	0.15	1.6	1.525	1.53
1LF05-18	9413	9402	1.96	5.19	29.2	517.9	0.80	5.93	0.07	0.48	76.1	27.1	100	1.55	582.0	9.15	13.30	0.15	2.0	0.970	1.57
1LF05-19	9413	9402	0.28	4.46	37.6	641.3	0.80	0.92	0.04	0.16	81.7	7.4	87	1.60	122.1	3.96	10.00	0.10	1.9	0.355	1.54
1LF05-20	9413	9402	0.16	1.95	405	30.5	0.25	0.40	13.00	0.08	13.20	65.4	711	0.60	47.4	5.76	4.75	0.10	0.4	0.040	0.03
1LF05-23	9413	9402	5.78	3.25	54.4	117.0	0.35	10.80	0.13	3.54	53.1	27.2	111	0.60	2560	9.95	8.75	0.25	1.0	4.52	0.44
1LF05-24	9413	9402	4.52	3.54	103.0	207.5	0.45	9.88	0.28	3.92	60.8	21.8	116	0.80	705.0	8.85	10.40	0.20	1.1	1.285	0.61
1LF05-25	9413	9402	3.44	4.29	73.2	292.5	0.45	8.35	0.13	3.54	64.4	15.9	113	1.00	577.0	7.30	11.70	0.15	1.1	0.905	0.80
1LF05-26	9413	9402	2.82	3.90	35.0	246.5	0.50	5.38	0.12	1.58	70.5	15.4	107	0.90	878.0	7.50	11.30	0.15	1.1	0.875	0.72
1LF05-27	9413	9402	1.76	6.48	11.0	710.8	1.10	6.57	0.07	0.82	72.6	11.2	87	2.25	295.0	6.08	17.95	0.15	1.9	1.330	2.07
1LF05-30	9413	9402	1.46	3.28	10.4	99.0	0.45	5.26	0.07	0.18	32.0	13.3	114	0.50	199.1	8.44	8.35	0.25	0.7	0.290	0.48
1LF06-01	9413	9402	4.20	3.08	177.5	62.0	0.35	11.75	0.32	1.96	58.0	47.9	109	0.95	1535.0	17.10	9.20	0.40	1.0	2.08	0.72
1LF06-07	9413	9402	2.26	6.76	6.4	220.5	1.45	9.66	0.37	0.76	68.5	40.1	94	1.80	1560.0	10.60	17.60	0.20	1.7	3.69	2.05
1LF06-08	9413	9402	2.18	9.22	48.6	101.0	1.85	11.90	0.14	0.88	151.5	33.2	87	2.40	991.0	17.55	24.10	0.45	3.4	1.520	2.67
1LF06-10	9413	9402	0.12	6.51	3.0	450.5	1.75	0.23	0.60	0.14	75.6	8.0	65	1.90	27.0	2.23	15.40	0.15	2.1	0.060	2.27
1LF06-11	9413	9402	0.10	8.26	5.8	469.5	2.40	0.52	0.41	0.02	95.6	11.5	74	3.10	22.2	2.94	21.05	0.20	2.9	0.085	3.30
1LF06-21	9413	9402	11.50	4.55	621	27.5	0.85	23.2	0.16	21.3	28.0	68.3	96	1.35	302.0	24.2	11.90	0.45	1.4	1.555	1.15

CERTIFICATION: [signature]

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Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
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Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm	Sn ppm	Sr ppm (ICP)	Ta ppm (ICP)	Te ppm (ICP)
A0203-43	413 402	47.0	35.4	1.20	575	1.10	5.26	4.7	54.9	680	17.0	104.0	0.002	2.24	0.05	4	2.6	251	0.45	0.05
A0203-44	413 402	34.0	30.8	1.05	915	22.95	4.20	2.6	52.8	2020	14.0	73.7	0.020	2.61	0.05	3	2.4	271	0.15	0.05
A0203-45	413 402	39.0	29.2	1.36	1220	38.10	2.53	3.7	57.1	1320	41.0	97.3	0.038	1.70	0.15	3	1.4	325	0.20	0.05
A0203-46	413 402	32.0	27.2	1.42	1215	19.95	0.43	3.6	60.5	1040	28.5	128.5	0.024	0.92	0.05	4	1.8	221	0.15	0.10
A0203-49	413 402	50.0	25.4	1.27	425	3.70	0.98	5.9	58.5	460	10.5	137.0	0.002	0.56	7.70	1	2.2	136.5	0.40	< 0.05
A0204-01	413 402	45.5	6.4	0.55	1225	38.20	4.68	2.5	199.0	620	50.5	36.5	0.028	4.99	0.25	15	2.0	194.0	0.15	0.25
A0204-02	413 402	32.5	4.2	0.68	1210	84.00	4.38	2.9	250	1040	68.5	30.3	0.030	5.36	0.45	13	1.6	228	0.10	0.25
A0204-03	413 402	36.0	15.2	1.41	715	8.95	0.33	2.7	175.0	360	31.5	176.5	0.006	7.45	0.20	17	3.0	177.5	0.10	0.65
A0204-48	413 402	30.0	13.2	1.08	780	36.30	2.86	2.1	61.0	1200	21.0	90.3	0.048	1.37	0.15	4	3.2	256	0.10	0.05
A0204-50	413 402	35.0	5.8	0.92	5780	14.80	0.05	2.5	203	6710	17.5	65.5	0.042	>10.00	2.65	23	0.8	160.5	0.15	1.00
A0204-52	413 402	22.5	10.4	1.02	1115	15.05	0.92	4.0	54.6	970	63.5	136.5	0.012	2.29	1.40	5	1.6	329	0.25	0.15
A0204-53	413 402	27.5	7.0	1.44	7240	4.35	0.08	2.1	60.6	880	32.0	68.6	0.006	>10.00	3.65	7	1.0	331	0.10	0.25
A0204-54	413 402	19.0	25.4	0.92	335	23.15	1.15	2.5	72.3	650	8.5	181.0	0.024	3.32	1.15	10	1.6	188.0	0.15	0.05
A0204-55	413 402	56.0	30.8	1.35	440	19.10	3.52	4.7	55.2	720	12.0	125.0	0.006	0.91	0.05	4	2.0	201	0.30	0.10
A9801-0201	413 402	39.0	22.8	1.02	405	12.75	1.11	1.9	55.3	680	17.5	103.5	0.012	0.74	0.10	4	1.4	457	0.15	0.05
A9801-0202	413 402	30.5	18.2	0.81	465	25.10	1.10	2.3	101.5	590	14.5	101.5	0.014	1.00	0.20	3	1.2	459	0.05	< 0.05
A9801-0203	413 402	47.5	22.6	1.06	590	31.30	1.83	3.8	131.5	850	87.5	143.5	0.012	1.16	0.15	3	2.4	462	0.20	0.10
A9801-0204	413 402	25.5	4.0	0.67	610	39.10	3.48	1.8	64.4	320	158.5	15.2	0.024	1.16	0.10	1	0.6	692	0.10	0.05
A9801-0205	413 402	19.5	5.2	0.43	590	51.06	2.18	1.8	64.1	290	236	30.1	0.020	1.05	0.15	2	1.4	417	0.05	0.20
A9801-0207	413 402	32.0	14.4	0.89	865	18.60	0.94	4.0	66.0	870	20.5	137.5	0.020	1.17	0.35	4	2.0	426	0.20	0.10
A9802-06	413 402	28.0	8.4	1.03	1080	22.00	2.32	2.1	54.2	870	203	68.0	0.020	0.94	0.20	4	1.2	712	0.10	0.25
1LF05-12	413 402	7.0	21.8	1.04	1605	6.90	0.16	0.6	66.8	990	486	5.7	0.006	7.37	1.05	13	29.8	24.4	< 0.05	0.35
1LF05-13	413 402	29.0	68.2	2.65	1590	6.60	0.20	0.9	54.6	260	256	17.6	0.002	4.17	0.45	5	16.8	23.4	< 0.05	0.05
1LF05-14	413 402	23.0	50.0	1.34	665	2.35	0.16	0.5	60.1	210	1080	4.5	0.004	9.00	0.70	15	17.4	19.6	< 0.05	0.80
1LF05-17	413 402	30.5	9.4	1.39	1655	1.25	0.40	0.7	18.0	130	45.0	71.4	< 0.002	0.43	0.45	1	18.2	54.4	< 0.05	< 0.05
1LF05-18	413 402	36.0	17.4	1.39	1535	0.85	0.38	0.7	18.6	130	198.0	78.2	< 0.002	1.64	0.40	1	17.2	46.0	< 0.05	< 0.05
1LF05-19	413 402	37.0	10.2	0.61	630	1.10	0.27	0.9	14.4	130	36.0	77.9	< 0.002	0.56	0.35	< 1	27.8	34.6	< 0.05	< 0.05
1LF05-20	413 402	6.0	41.0	8.56	2280	0.55	< 0.01	0.6	768	390	19.0	1.9	< 0.002	0.44	1.40	< 1	0.6	435	< 0.05	< 0.05
1LF05-23	413 402	25.5	44.0	1.95	1370	2.55	0.47	0.7	37.6	300	377	20.8	0.006	3.62	0.40	4	20.8	53.8	< 0.05	0.05
1LF05-24	413 402	28.5	122.0	2.02	1515	0.95	0.45	0.7	36.0	580	492	30.3	< 0.002	2.71	1.05	3	18.6	83.0	< 0.05	0.05
1LF05-25	413 402	30.0	35.4	1.74	1150	1.40	0.40	0.8	23.6	200	496	43.5	< 0.002	2.16	1.00	2	23.2	50.2	< 0.05	0.05
1LF05-26	413 402	32.0	21.4	1.76	1210	0.90	0.58	0.8	22.0	390	191.0	38.2	< 0.002	1.18	0.50	1	18.4	67.8	< 0.05	< 0.05
1LF05-27	413 402	35.5	15.0	1.55	1275	1.15	0.52	1.3	26.6	230	147.5	109.5	< 0.002	0.69	0.25	1	45.0	70.0	0.05	< 0.05
1LF05-30	413 402	14.5	32.2	2.33	1705	2.70	0.29	0.5	23.4	80	128.5	23.6	< 0.002	1.55	0.15	1	7.4	37.8	< 0.05	0.05
1LF06-01	413 402	27.0	22.4	1.92	2830	2.05	0.15	0.7	53.3	330	324	41.7	< 0.002	9.43	0.95	7	18.2	24.0	< 0.05	< 0.05
1LF06-07	413 402	31.5	17.8	2.39	2050	1.35	0.71	0.7	21.0	190	61.0	95.7	< 0.002	2.66	0.35	3	35.4	93.2	< 0.05	0.10
1LF06-08	413 402	64.5	28.8	3.49	4150	1.95	1.01	0.9	26.6	250	114.5	131.0	0.002	6.40	0.40	6	49.0	129.5	< 0.05	0.05
1LF06-10	413 402	37.0	11.4	0.78	385	1.45	0.44	1.4	24.4	370	28.5	106.5	0.002	0.21	0.15	< 1	2.2	70.9	< 0.05	< 0.05
1LF06-11	413 402	46.0	19.4	1.24	320	1.60	0.33	1.7	28.8	1180	14.5	159.5	0.004	1.52	0.85	1	3.8	55.7	0.05	< 0.05
1LF06-21	413 402	12.0	67.0	4.52	3240	4.65	0.52	1.0	93.6	200	1415	60.1	0.002	>10.00	5.15	4	18.8	76.2	< 0.05	0.35

CERTIFICATION:

ALS Chemex

Aurora Laboratory Services Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver

British Columbia, Canada V7J 2C1

PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE		Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
A0203-43	9413	9402	14.2	0.14	0.60	3.3	267	2.5	9.6	176	29.0	< 10
A0203-44	9413	9402	11.4	0.10	0.38	6.5	348	1.9	14.6	104	25.0	< 10
A0203-45	9413	9402	10.4	0.15	0.60	4.8	201	0.8	12.5	172	34.0	< 10
A0203-46	9413	9402	8.6	0.19	0.72	3.2	185	1.1	11.0	132	46.5	< 10
A0203-49	9413	9402	15.8	0.16	0.66	1.9	120	0.9	16.8	106	20.0	< 10
A0204-01	9413	9402	12.4	0.06	0.28	14.1	357	3.1	10.4	28	47.0	< 10
A0204-02	9413	9402	8.2	0.05	0.22	10.2	213	3.2	11.5	206	35.5	< 10
A0204-03	9413	9402	12.2	0.12	1.02	1.7	118	3.3	7.8	360	15.5	< 10
A0204-48	9413	9402	10.0	0.11	0.46	5.1	243	4.0	10.0	48	24.0	< 10
A0204-50	9413	9402	5.0	0.04	0.72	6.9	328	0.5	46.4	234	15.5	< 10
A0204-52	9413	9402	7.6	0.12	0.92	4.7	260	2.9	11.7	278	43.0	< 10
A0204-53	9413	9402	3.8	0.04	0.50	2.1	117	2.1	22.5	70	19.0	< 10
A0204-54	9413	9402	6.0	0.15	1.26	9.7	666	1.5	8.1	732	24.0	< 10
A0204-55	9413	9402	18.6	0.15	0.70	2.4	242	0.6	9.5	172	32.0	< 10
A9801-0201	9413	9402	12.8	0.10	0.84	5.5	311	0.8	16.5	174	16.0	< 10
A9801-0202	9413	9402	9.6	0.09	0.82	5.2	298	1.0	9.8	238	11.5	< 10
A9801-0203	9413	9402	15.6	0.13	1.02	3.8	205	2.0	12.8	126	17.5	< 10
A9801-0204	9413	9402	7.2	0.06	0.12	6.6	268	1.0	10.2	194	10.0	< 10
A9801-0205	9413	9402	6.0	0.05	0.22	5.6	109	1.8	7.1	42	10.5	< 10
A9801-0207	9413	9402	10.0	0.12	0.98	8.1	407	1.8	10.0	58	18.5	< 10
A9802-06	9413	9402	8.6	0.08	0.50	9.3	346	1.0	11.6	586	19.5	< 10
1LF05-12	9413	9402	3.2	< 0.01	0.12	5.5	74	1.0	19.8	1050	31.5	70
1LF05-13	9413	9402	9.2	0.02	0.16	2.7	46	1.4	7.5	988	44.0	40
1LF05-14	9413	9402	6.0	< 0.01	0.08	9.5	102	0.6	11.1	4290	46.0	350
1LF05-17	9413	9402	14.2	0.04	0.48	2.0	33	1.1	6.0	354	55.0	10
1LF05-18	9413	9402	17.2	0.04	0.52	2.4	38	1.0	5.3	232	69.0	10
1LF05-19	9413	9402	19.0	0.04	0.50	2.4	29	1.3	5.3	94	63.5	< 10
1LF05-20	9413	9402	0.8	0.03	0.04	1.4	76	0.3	4.3	82	17.0	10
1LF05-23	9413	9402	8.2	0.02	0.20	3.8	70	1.9	8.9	1125	38.0	50
1LF05-24	9413	9402	9.2	0.02	0.26	2.7	38	1.6	6.6	1220	38.5	60
1LF05-25	9413	9402	9.8	0.03	0.34	2.2	41	1.7	7.3	1250	42.5	50
1LF05-26	9413	9402	10.4	0.03	0.28	1.9	37	1.5	6.3	512	41.0	20
1LF05-27	9413	9402	16.0	0.05	0.72	2.5	53	1.7	6.0	386	69.5	10
1LF05-30	9413	9402	7.2	0.01	0.22	0.8	13	0.9	5.8	176	24.0	10
1LF06-01	9413	9402	7.0	0.01	0.30	2.3	36	1.2	9.4	618	37.0	20
1LF06-07	9413	9402	12.2	0.04	0.64	2.3	37	1.4	5.5	380	58.5	10
1LF06-08	9413	9402	25.2	0.05	0.82	3.9	52	2.4	9.0	476	114.5	10
1LF06-10	9413	9402	16.8	0.07	0.68	2.9	49	0.8	6.8	46	69.5	< 10
1LF06-11	9413	9402	19.8	0.09	1.08	5.7	73	1.3	10.8	50	106.5	< 10
1LF06-21	9413	9402	8.6	0.02	0.86	1.2	46	1.8	3.9	5120	37.0	230

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE		Weight Kg	Au ppb ICP	Pt ppb ICP	Pd ppb ICP	Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %
1LF06-22	9413	9402	0.50	4	< 5	< 2	10.33	0.08	0.63	0.02	10.31	2.21	3.64	0.23	0.49	0.10	63.39	< 0.01	0.45	7.89	99.77
2LF01-31	9413	9402	1.52	< 2	< 5	< 2	14.64	0.12	4.95	0.01	7.28	4.02	4.70	0.31	0.54	0.07	50.44	0.01	0.52	11.70	99.31
2LF01-44	9413	9402	0.48	22	< 5	2	12.27	0.10	0.40	< 0.01	21.93	3.40	6.03	0.41	1.48	0.10	32.61	< 0.01	0.55	18.77	98.05
2LF05-16	9413	9402	0.96	34	< 5	2	5.39	0.02	0.21	0.01	13.66	0.59	3.06	0.20	0.63	0.11	64.73	< 0.01	0.28	10.24	99.13
2LF05-34	9413	9402	0.62	10	< 5	4	3.67	0.01	0.14	0.02	9.16	0.44	2.24	0.15	0.36	0.05	72.63	< 0.01	0.19	8.94	98.00
2LF05-38	9413	9402	0.26	8	< 5	< 2	11.16	0.06	0.11	0.01	16.11	1.41	4.76	0.18	0.47	0.06	55.16	< 0.01	0.56	8.83	98.88
2LF05-42	9413	9402	0.42	64	< 5	2	6.74	0.04	0.17	< 0.01	28.12	0.99	5.03	0.30	0.66	0.03	38.12	< 0.01	0.29	19.04	99.53
2LF06-02	9413	9402	0.84	60	< 5	< 2	7.20	0.03	2.63	< 0.01	14.79	0.75	5.66	0.29	0.70	0.09	53.75	< 0.01	0.41	12.90	99.20
2LF06-04	9413	9402	2.02	60	< 5	< 2	8.61	0.07	0.11	0.01	13.93	1.68	1.69	0.15	0.51	0.03	61.59	< 0.01	0.45	9.26	98.09
2LF06-35	9413	9402	1.46	< 2	< 5	< 2	16.05	0.13	0.32	0.01	7.72	3.74	3.20	0.15	0.82	0.07	58.09	< 0.01	0.65	8.38	99.33
2LF06-40	9413	9402	1.70	2	< 5	< 2	16.63	0.14	0.26	0.01	8.98	4.23	2.39	0.16	0.75	0.07	57.30	< 0.01	0.67	8.25	99.84
3LF01-32	9413	9402	0.18	14	< 5	< 2	6.85	0.04	4.27	0.01	15.01	1.75	6.44	0.51	1.85	0.11	46.23	0.01	0.30	14.71	98.09
3LF05-15	9413	9402	0.16	726	5	4	3.16	< 0.01	0.12	0.01	21.82	0.41	2.01	0.13	0.27	0.06	56.87	< 0.01	0.17	13.17	98.20
3LF05-19A	9413	9402	0.16	16	< 5	< 2	5.66	0.04	0.18	0.01	12.75	1.19	1.40	0.13	0.34	0.05	68.21	< 0.01	0.28	7.94	98.18
3LF05-33	9413	9402	0.88	174	< 5	2	5.05	0.03	0.14	0.01	16.94	0.65	2.78	0.19	0.40	0.06	60.23	< 0.01	0.26	11.57	98.31
3LF05-39	9413	9402	0.70	64	< 5	2	6.86	0.01	0.25	< 0.01	32.67	0.47	6.29	0.42	0.28	0.13	32.82	< 0.01	0.37	19.02	99.59
3LF05-43	9413	9402	0.22	38	< 5	< 2	9.31	0.04	0.21	< 0.01	30.01	1.24	6.12	0.30	0.87	0.03	30.49	< 0.01	0.32	20.83	99.77
3LF06-03	9413	9402	0.18	280	< 5	< 2	6.43	0.04	2.26	< 0.01	29.98	0.86	5.39	0.25	0.51	0.11	30.47	< 0.01	0.38	21.32	98.00
3LF06-05	9413	9402	0.08	584	< 5	< 2	4.37	0.03	0.15	0.01	29.20	0.83	1.93	0.16	0.08	0.03	46.61	< 0.01	0.20	16.27	99.87
3LF06-06	9413	9402	0.26	462	< 5	< 2	4.41	0.04	0.13	< 0.01	26.55	0.86	1.82	0.16	0.23	0.04	49.12	< 0.01	0.26	15.19	98.81
3LF06-36	9413	9402	0.08	190	< 10	< 4	3.66	0.01	0.13	0.03	18.10	0.63	3.69	0.28	0.62	0.03	58.36	< 0.01	0.12	12.82	98.48
3LF06-41	9413	9402	0.46	6	< 5	< 2	12.70	0.12	0.25	0.01	13.39	3.28	2.26	0.15	0.53	0.06	56.12	0.03	0.65	10.11	99.66
4LF06-37	9413	9402	0.48	22	< 5	< 2	10.88	0.09	0.15	0.01	18.40	2.44	3.53	0.25	0.56	0.04	49.27	0.01	0.55	13.36	99.54

CERTIFICATION: [illegible]

ALS Chemex

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British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
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CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Ag ppm	Ba ppm	Ce ppm	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm
1LF06-22	94139402	< 1	637	63.5	16.5	230	1.9	100	3.1	1.8	0.6	15	4.4	5	0.6	31.0	0.3	6	9	26.5
2LF01-31	94139402	< 1	1085	109.5	16.0	160	3.2	55	4.9	2.9	1.3	21	7.3	3	1.1	54.0	0.4	4	20	45.5
2LF01-44	94139402	10	847	82.0	64.0	170	2.4	1730	4.4	2.6	1.0	18	6.1	3	1.0	42.0	0.4	10	15	34.5
2LF05-16	94139402	4	137.0	49.5	28.5	180	0.6	1700	3.0	1.8	0.4	9	3.7	1	0.6	24.0	0.2	6	5	20.5
2LF05-34	94139402	2	138.5	41.0	22.0	350	0.7	750	3.8	2.4	0.4	8	4.1	< 1	0.9	21.0	0.3	10	4	18.0
2LF05-38	94139402	1	511	69.0	28.0	180	1.5	580	3.6	2.2	0.5	19	4.7	5	0.8	34.5	0.3	6	11	29.5
2LF05-42	94139402	5	223	12.5	56.5	150	1.0	1365	1.5	1.0	0.1	13	1.4	1	0.4	7.5	0.1	6	5	5.5
2LF06-02	94139402	7	201	66.5	37.0	140	0.7	2120	2.5	1.4	0.5	12	3.9	2	0.5	30.0	0.2	6	10	26.0
2LF06-04	94139402	1	557	73.5	45.5	180	1.4	1330	2.5	1.6	0.6	13	4.1	7	0.6	33.0	0.2	6	8	27.5
2LF06-35	94139402	< 1	1175	114.0	14.5	190	3.0	85	4.8	2.9	1.3	24	7.4	6	1.0	54.5	0.4	6	14	47.0
2LF06-40	94139402	< 1	1305	118.0	17.0	200	3.2	170	4.6	2.9	1.4	24	7.3	7	1.0	55.0	0.4	6	15	48.0
3LF01-32	94139402	9	431	57.0	27.0	200	1.2	2140	4.7	2.4	1.3	10	6.4	1	1.0	28.0	0.3	6	9	26.0
3LF05-15	94139402	13	103.0	32.5	67.5	250	0.4	7940	1.9	1.2	0.2	6	2.6	< 1	0.4	15.0	0.1	8	3	14.0
3LF05-19A	94139402	3	415	56.0	53.0	240	1.0	330	2.0	1.2	0.5	8	3.3	6	0.4	25.0	0.1	6	5	20.0
3LF05-33	94139402	12	187.0	56.5	48.5	170	0.7	6570	3.4	2.1	0.3	10	4.2	1	0.8	26.0	0.3	8	5	23.0
3LF05-39	94139402	5	140.0	55.5	78.5	160	0.5	3780	3.3	1.9	0.5	12	4.3	2	0.7	27.0	0.2	8	6	24.0
3LF05-43	94139402	20	263	8.5	84.0	80	1.1	3200	1.4	0.9	0.1	17	1.3	1	0.3	6.0	0.1	6	6	4.0
3LF06-03	94139402	78	192.5	39.5	154.0	170	0.6	>10000	1.6	0.9	0.3	12	2.7	1	0.3	18.0	0.1	8	7	17.0
3LF06-05	94139402	13	216	30.0	189.5	270	0.7	>10000	1.3	0.8	0.3	7	2.1	3	0.3	14.5	0.1	8	3	11.5
3LF06-06	94139402	11	251	42.5	149.0	160	0.7	6860	1.6	0.9	0.4	8	2.7	4	0.3	20.0	0.1	6	4	16.0
3LF06-36	94139402	6	144.0	15.5	48.5	400	0.5	2840	1.1	0.7	0.2	6	1.3	1	0.2	8.5	< 0.1	8	2	6.0
3LF06-41	94139402	2	989	95.5	42.0	200	2.4	420	3.7	2.3	1.2	18	6.0	7	0.8	42.0	0.3	8	12	37.5
4LF06-37	94139402	3	707	84.5	29.5	200	1.8	1555	3.1	2.1	1.0	17	4.9	7	0.7	39.0	0.3	8	10	33.0

CERTIFICATION: [signature]

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SAMPLE	PREP CODE	Ni ppm	Pb ppm	Pr ppm	Rb ppm	Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
1LF06-22	94139402	35	250	7.3	97.6	4.6	32	64.8	0.5	0.6	12	< 0.5	0.3	2.0	40	10	15.0	1.8	205	183.5
2LF01-31	94139402	35	100	12.7	167.5	7.8	35	127.5	1.5	1.0	16	0.5	0.4	2.0	55	7	25.0	2.7	425	113.5
2LF01-44	94139402	65	660	9.8	132.0	6.2	114	65.6	1.0	0.9	11	0.5	0.4	3.0	130	11	24.0	2.4	>10000	100.0
2LF05-16	94139402	30	210	5.7	25.8	3.8	54	46.3	< 0.5	0.6	6	< 0.5	0.3	2.5	60	11	18.0	1.6	2590	49.5
2LF05-34	94139402	45	55	4.9	23.2	3.6	15	38.0	< 0.5	0.7	5	< 0.5	0.4	4.5	120	14	26.5	2.1	220	41.0
2LF05-38	94139402	30	105	8.0	66.0	4.9	55	37.9	0.5	0.6	11	< 0.5	0.3	1.5	55	14	20.0	1.9	2180	160.0
2LF05-42	94139402	30	580	1.4	42.6	1.3	31	60.5	< 0.5	0.3	6	< 0.5	0.1	0.5	35	12	8.5	1.0	700	51.0
2LF06-02	94139402	35	435	6.9	31.8	4.2	37	110.0	0.5	0.5	8	< 0.5	0.2	1.5	45	16	12.0	1.4	1290	69.0
2LF06-04	94139402	20	100	7.7	71.0	4.7	42	64.1	0.5	0.5	14	< 0.5	0.2	2.0	35	7	13.5	1.5	225	238
2LF06-35	94139402	35	35	13.1	158.0	8.1	52	104.0	1.0	1.0	18	0.5	0.4	2.5	85	9	25.0	2.6	230	198.0
2LF06-40	94139402	40	135	13.2	171.0	7.8	41	72.4	1.0	0.9	19	0.5	0.4	2.5	75	8	24.5	2.6	410	234
3LF01-32	94139402	30	870	6.8	68.0	6.0	70	116.5	0.5	1.0	7	< 0.5	0.3	1.5	30	11	24.5	2.0	>10000	67.0
3LF05-15	94139402	20	270	3.6	17.0	2.5	168	30.9	< 0.5	0.3	3	< 0.5	0.1	1.5	40	8	11.5	1.1	3830	30.0
3LF05-19A	94139402	45	455	5.7	50.4	3.5	33	27.9	< 0.5	0.4	11	< 0.5	0.1	2.5	20	5	11.0	1.1	420	216
3LF05-33	94139402	35	440	6.1	28.2	4.3	90	44.5	< 0.5	0.6	5	< 0.5	0.3	2.0	60	15	21.5	1.8	2190	46.0
3LF05-39	94139402	30	170	6.5	18.8	4.4	56	20.3	< 0.5	0.6	8	< 0.5	0.3	2.0	35	14	19.0	1.7	3720	79.0
3LF05-43	94139402	40	1265	1.1	50.2	1.0	49	84.2	< 0.5	0.2	6	< 0.5	0.1	1.0	40	14	7.5	1.0	2430	46.5
3LF06-03	94139402	45	4350	4.5	30.2	3.1	322	96.8	0.5	0.3	6	< 0.5	0.1	1.5	35	16	8.0	0.8	9480	52.0
3LF06-05	94139402	25	260	3.2	28.8	2.0	59	36.8	< 0.5	0.3	6	< 0.5	0.1	0.5	10	4	7.0	0.7	780	94.5
3LF06-06	94139402	30	205	4.6	33.8	2.9	77	37.1	< 0.5	0.3	8	< 0.5	0.1	1.0	15	4	8.5	0.8	1125	135.5
3LF06-36	94139402	35	290	1.7	26.0	1.2	54	31.6	< 0.5	0.1	4	< 0.5	0.1	0.5	5	6	6.5	0.7	7810	69.0
3LF06-41	94139402	55	220	10.2	135.0	6.5	37	61.4	0.5	0.8	17	0.5	0.3	2.5	60	8	20.5	2.0	390	253
4LF06-37	94139402	40	180	9.2	96.0	5.5	49	78.9	0.5	0.6	15	< 0.5	0.3	2.0	50	10	17.0	1.9	565	243

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE		Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)	Ge ppm (ICP)	Hf ppm	In ppm	K % (ICP)
1LF06-22	9413	9402	0.52	5.68	741	656.4	0.95	0.66	0.44	0.30	56.0	15.4	143	1.90	107.3	7.28	15.80	0.30	2.3	0.120	1.86
2LF01-31	9413	9402	0.48	7.68	59.8	1068.0	1.70	1.46	3.60	1.20	93.0	15.0	75	3.15	38.4	4.88	20.50	0.30	3.0	0.070	3.32
2LF01-44	9413	9402	11.50	6.39	183.0	45.5	1.20	20.7	0.29	90.8	62.0	59.8	84	2.30	1950.0	14.45	17.40	0.55	2.0	6.65	2.47
2LF05-16	9413	9402	5.54	2.64	177.5	124.5	0.30	9.95	0.15	11.55	45.4	26.2	88	0.65	1885.0	9.35	9.25	0.35	0.9	2.02	0.44
2LF05-34	9413	9402	2.88	2.31	46.4	152.0	0.35	5.90	0.12	0.76	39.8	24.9	230	0.70	815.0	7.90	7.75	0.30	0.9	0.795	0.44
2LF05-38	9413	9402	1.60	6.39	75.6	106.0	0.75	5.90	0.08	9.44	64.9	28.5	100	1.60	666.0	11.45	20.25	0.45	1.9	2.53	1.18
2LF05-42	9413	9402	8.04	3.95	204	34.0	0.45	17.20	0.12	2.82	11.00	51.3	100	1.00	1725.0	19.85	13.40	0.60	0.8	1.860	0.77
2LF06-02	9413	9402	11.25	3.75	128.5	199.5	0.40	20.5	1.75	4.38	61.6	38.9	77	0.70	2460	9.63	12.80	0.40	1.5	3.35	0.58
2LF06-04	9413	9402	2.46	4.65	39.0	195.0	0.75	3.66	0.06	0.94	70.4	44.7	129	1.55	1480.0	9.93	14.65	0.40	2.1	1.720	1.46
2LF06-35	9413	9402	0.26	8.28	58.4	1296.0	1.80	0.78	0.23	0.46	102.0	13.4	87	3.05	89.7	5.21	24.45	0.30	2.3	0.305	2.93
2LF06-40	9413	9402	0.68	8.26	20.6	1037.5	2.10	1.84	0.17	1.26	102.0	16.5	96	3.30	137.4	5.98	25.20	0.35	2.3	0.310	3.24
3LF01-32	9413	9402	11.45	3.63	63.8	291.5	0.85	17.60	3.00	116.5	52.2	27.6	124	1.35	2450	9.97	10.40	0.40	1.3	5.16	1.41
3LF05-15	9413	9402	15.65	1.81	500	82.0	0.25	16.90	0.09	20.0	27.6	60.8	135	0.45	8950	14.60	6.20	0.50	0.6	10.80	0.29
3LF05-19A	9413	9402	4.52	3.27	17.4	50.5	0.65	15.15	0.14	1.82	51.2	54.4	139	1.15	364.0	9.58	8.95	0.35	1.7	0.810	1.04
3LF05-33	9413	9402	14.90	2.64	252	47.5	0.35	15.05	0.09	9.60	49.8	55.7	84	0.75	7220	11.70	9.60	0.45	0.8	10.00	0.50
3LF05-39	9413	9402	6.44	4.08	208	54.0	0.25	18.45	0.17	15.60	49.0	83.4	86	0.55	4650	22.5	12.70	0.75	1.3	6.25	0.34
3LF05-43	9413	9402	24.6	5.08	109.5	148.5	0.60	49.1	0.14	12.80	6.78	86.4	62	1.20	3970	20.5	18.40	0.60	0.8	5.60	0.90
3LF06-03	9413	9402	86.7	3.47	1945	33.0	0.35	173.0	1.45	32.0	37.1	147.6	78	0.75	>10000	20.4	12.80	0.80	1.1	34.4	0.54
3LF06-05	9413	9402	15.35	2.36	209	42.5	0.30	17.30	0.08	3.98	26.4	183.0	156	0.70	>10000	19.75	7.15	0.65	0.9	4.85	0.55
3LF06-06	9413	9402	13.45	2.42	178.0	28.0	0.30	17.05	0.08	5.56	32.0	127.9	66	0.70	7840	17.25	7.60	0.60	1.1	8.06	0.62
3LF06-36	9413	9402	8.74	2.06	118.5	150.5	0.35	16.55	0.08	33.3	14.00	47.6	212	0.55	3390	12.20	5.85	0.35	0.7	9.04	0.49
3LF06-41	9413	9402	3.20	7.12	24.4	83.5	1.65	7.13	0.20	2.14	82.7	43.7	105	2.65	512.0	9.91	20.15	0.40	2.0	0.675	2.75
4LF06-37	9413	9402	4.54	5.86	120.0	194.0	1.20	12.15	0.10	1.74	78.9	28.8	105	2.00	1755.0	12.45	18.00	0.50	2.2	1.885	1.94

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 18-JUL-2002
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm	Sn ppm	Sr ppm (ICP)	Ta ppm (ICP)	Te ppm (ICP)
1LF06-22	9413 9402	30.0	25.4	2.22	1670	1.45	0.19	1.7	37.4	440	308	88.8	< 0.002	0.32	2.30	< 1	18.2	57.8	0.10	< 0.05
2LF01-31	9413 9402	52.5	10.0	2.80	2280	0.40	0.30	5.1	33.0	290	130.5	142.0	< 0.002	0.31	0.80	1	22.0	122.0	0.35	< 0.05
2LF01-44	9413 9402	29.0	8.4	3.18	3320	4.45	0.30	2.4	69.3	400	688	121.0	0.006	6.23	1.00	12	56.2	53.8	0.15	0.30
2LF05-16	9413 9402	22.5	14.2	1.69	1385	1.15	0.35	1.0	31.6	430	278	25.4	< 0.002	4.56	0.75	6	20.2	40.4	0.05	0.05
2LF05-34	9413 9402	22.0	7.4	1.60	1310	5.70	0.29	1.2	51.5	280	94.0	23.3	0.016	1.14	0.60	3	11.4	37.6	0.05	0.10
2LF05-38	9413 9402	30.5	67.2	2.83	1420	0.90	0.27	1.5	32.2	260	133.0	64.9	< 0.002	3.66	1.40	4	33.8	34.2	0.05	< 0.05
2LF05-42	9413 9402	5.5	14.4	2.79	2630	1.95	0.47	1.1	34.4	130	570	43.9	< 0.002	6.11	1.40	6	19.2	59.7	0.05	0.15
2LF06-02	9413 9402	31.0	29.4	3.15	2030	1.40	0.41	1.8	36.4	400	654	31.3	< 0.002	4.11	2.10	6	24.8	108.5	0.05	0.05
2LF06-04	9413 9402	33.0	10.8	0.99	1060	2.45	0.42	2.3	26.6	110	90.0	70.8	< 0.002	2.51	0.60	4	29.6	62.6	0.10	0.05
2LF06-35	9413 9402	53.5	17.6	1.82	1105	0.70	0.54	3.4	35.6	340	49.0	142.0	< 0.002	0.23	1.60	< 1	34.4	94.4	0.25	< 0.05
2LF06-40	9413 9402	51.0	14.2	1.32	1060	0.85	0.43	2.8	35.6	240	136.0	158.5	< 0.002	0.50	0.40	1	27.0	65.4	0.20	< 0.05
3LF01-32	9413 9402	28.0	5.8	3.69	3700	1.15	0.19	1.7	31.4	500	1020	67.8	< 0.002	2.56	0.80	11	40.4	118.5	0.05	0.25
3LF05-15	9413 9402	13.0	10.2	1.07	835	2.50	0.22	0.8	19.6	200	289	16.6	< 0.002	>10.00	0.90	20	46.0	28.4	< 0.05	0.20
3LF05-19A	9413 9402	22.5	10.2	0.85	920	2.65	0.23	0.8	50.8	150	500	51.7	< 0.002	4.25	0.75	5	23.4	27.4	< 0.05	0.15
3LF05-33	9413 9402	23.5	10.6	1.55	1355	2.00	0.36	0.7	37.4	240	578	27.6	0.002	5.59	0.70	12	38.8	40.6	< 0.05	0.10
3LF05-39	9413 9402	25.0	71.8	3.35	3240	1.65	0.12	1.1	51.1	510	388	20.2	< 0.002	>10.00	2.95	13	29.8	20.0	0.05	0.10
3LF05-43	9413 9402	3.5	31.6	3.15	2700	1.50	0.65	1.0	48.6	150	1780	51.2	< 0.002	3.73	0.90	9	34.4	81.6	< 0.05	0.10
3LF06-03	9413 9402	17.5	22.0	2.81	1725	1.30	0.38	0.9	55.9	440	5380	31.6	< 0.002	>10.00	13.20	35	102.5	92.8	< 0.05	0.55
3LF06-05	9413 9402	12.0	6.8	0.98	1125	2.80	0.22	1.2	28.4	90	325	29.7	< 0.002	>10.00	4.15	21	42.0	32.0	< 0.05	0.10
3LF06-06	9413 9402	15.5	6.4	0.92	1035	1.00	0.22	1.1	25.4	90	219	32.1	< 0.002	>10.00	2.80	23	39.6	32.2	< 0.05	0.05
3LF06-36	9413 9402	7.5	6.6	2.02	2080	2.75	0.22	0.9	35.6	90	298	25.8	< 0.002	4.89	1.25	7	22.8	32.0	< 0.05	0.25
3LF06-41	9413 9402	36.0	10.6	1.35	1375	1.00	0.37	1.5	62.7	250	321	133.0	< 0.002	3.59	0.45	4	25.6	58.4	0.05	0.20
4LF06-37	9413 9402	39.5	10.0	1.95	2200	1.10	0.47	1.3	29.6	190	201	97.7	< 0.002	2.56	0.55	3	28.4	79.8	0.05	0.05

CERTIFICATION: [signature]

ALS Chemex
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Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : FRANK AND ACE
Comments: ATTN: LOUIS E. DOYLE

Page Number :3-F
Total Pages :3
Certificate Date: 18-JUL-200
Invoice No. : I0219395
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0219395

SAMPLE	PREP CODE	Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
1LF06-22	9413 9402	18.4	0.06	0.60	2.8	45	1.8	5.5	236	58.5	10
2LF01-31	9413 9402	22.8	0.10	1.12	2.6	48	1.9	9.3	476	86.0	30
2LF01-44	9413 9402	13.0	0.06	0.94	3.8	124	2.3	6.8	>10000	63.0	2040
2LF05-16	9413 9402	7.8	0.03	0.22	2.9	41	1.8	10.5	2850	31.5	120
2LF05-34	9413 9402	6.0	0.03	0.20	6.2	109	3.7	13.4	254	34.5	10
2LF05-38	9413 9402	14.8	0.05	0.52	2.3	50	2.0	6.3	2500	60.5	60
2LF05-42	9413 9402	7.4	0.02	0.36	1.0	30	2.4	3.6	748	26.5	30
2LF06-02	9413 9402	11.4	0.04	0.24	2.1	36	2.5	7.2	1380	48.5	60
2LF06-04	9413 9402	16.8	0.05	0.46	2.2	34	2.2	6.0	236	64.5	10
2LF06-35	9413 9402	22.0	0.10	0.86	2.6	68	2.9	7.2	228	72.5	10
2LF06-40	9413 9402	22.4	0.10	0.98	3.2	64	2.6	7.2	410	76.0	10
3LF01-32	9413 9402	10.0	0.04	0.52	1.8	26	1.3	12.3	>10000	45.5	2350
3LF05-15	9413 9402	4.4	0.01	0.16	1.9	35	1.9	7.8	4080	20.5	180
3LF05-19A	9413 9402	10.6	0.03	0.40	3.2	18	1.3	6.5	484	51.0	30
3LF05-33	9413 9402	7.0	0.01	0.26	2.6	48	2.6	10.0	2380	32.0	100
3LF05-39	9413 9402	11.4	0.02	0.28	2.4	28	1.6	6.9	4000	46.0	140
3LF05-43	9413 9402	8.8	0.02	0.44	1.4	36	2.1	3.5	2580	31.0	100
3LF06-03	9413 9402	8.2	0.01	0.32	1.8	27	1.6	6.3	9750	38.5	400
3LF06-05	9413 9402	6.8	0.02	0.22	0.9	14	1.5	5.4	842	29.0	40
3LF06-06	9413 9402	7.8	0.01	0.22	1.2	14	0.9	4.8	1150	34.0	50
3LF06-36	9413 9402	5.0	0.01	0.20	0.7	13	1.5	4.1	7880	22.5	350
3LF06-41	9413 9402	17.6	0.06	0.82	2.8	52	2.1	6.9	468	69.0	10
4LF06-37	9413 9402	20.0	0.04	0.56	2.5	40	1.8	6.6	618	74.5	30

CERTIFICATION: [signature]

Eco Tech LABORATORY LTD.

ASSAYING
GEOCHEMISTRY
ANALYTICAL CHEMISTRY
ENVIRONMENTAL TESTING

10041 Dallas Drive, Kamloops, B.C. V2C 6T4
Phone (250) 573-5700 Fax (250) 573-4557
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CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

24-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 24
Sample Type: Core
Project #: ACE
Drill Hole #: ACE - 02-04
Samples submitted by: Chris Wild

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
1	8199	<1	0.32	<5	0.2	<0.02	0.3
2	8200	<1	0.18	<5	0.3	<0.02	0.2
3	8201	<1	0.05	<5	0.7	0.02	0.1
4	8202	2	0.24	<5	1.6	0.08	0.5
5	8203	1	0.11	<5	1.1	0.07	0.2
6	8204	<1	0.04	<5	1.4	0.07	0.1
7	8205	1	0.09	<5	1.8	0.08	0.2
8	8206	3	0.07	<5	4.1	0.12	0.2
9	8207	2	0.06	<5	6.1	0.20	0.2
10	8208	2	0.07	7	9.7	0.16	0.2
11	8209	2	0.07	<5	10.1	0.08	0.2
12	8210	1	0.03	<5	5.8	0.10	0.2
13	8211	<1	0.05	6	6.9	0.12	0.2
14	8218	<1	0.03	<5	3.2	0.03	0.2
15	8219	2	0.15	<5	1.4	0.07	0.2
16	8220	<1	0.04	<5	2.6	0.07	0.3
17	8221	<1	0.04	<5	2.6	0.07	0.3
18	8222	<1	0.09	<5	3.4	0.07	0.3
19	8223	3	0.09	<5	3.3	0.04	0.4
20	8224	2	0.12	<5	2.8	0.10	0.7
21	8225	<1	0.36	<5	1.4	0.08	0.7
22	8226	<1	0.06	<5	2.7	0.09	0.2
23	8227	<1	0.04	<5	1.0	0.03	0.1
24	8228	<1	0.16	<5	1.2	0.03	0.2

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
QC DATA:							
Resplit:							
1	8199	<1	0.33	<5	0.2	<0.02	0.4
Repeat:							
10	8208	2	0.07	7	9.8	0.13	0.2
Standard:							
STANDARD		2	1.20	242	1.4	1.00	6.7

NOTE: Results run by ACME Analytical Laboratories Ltd.

XLS/02
E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

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Phone (250) 573-5700 Fax (250) 573-4557
email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

24-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 65
Sample Type: Core
Project #: Frank Creek
Drill Hole #: FC-02-01 to FC-02-06
Samples submitted by: Chris Wild

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
25	8240	<1	0.05	129	1.4	0.03	2.8
26	8241	<1	0.07	1702	11.7	0.25	10.6
[illegible]7	8242	<1	0.05	183	1.3	0.04	2.3
[illegible]	8243	<1	0.11	2820	9.2	0.24	15.4
29	8244	<1	0.10	35	0.6	0.02	9.0
30	8245	<1	0.04	35	0.6	<0.02	0.9
31	8246	<1	0.02	<5	0.2	<0.02	0.2
32	8247	<1	<0.02	5	0.3	<0.02	<0.1
33	8248	<1	0.02	<5	0.3	<0.02	0.1
34	8249	<1	0.02	<5	0.4	<0.02	0.1
35	8250	<1	<0.02	<5	0.2	<0.02	0.1
36	8251	1	0.16	38	2.3	0.02	0.3
37	8252	<1	0.03	<5	<0.1	<0.02	0.1
38	8253	<1	0.02	<5	0.3	<0.02	<0.1
39	8254	<1	0.04	<5	<0.1	<0.02	0.2
40	8255	<1	15.19	6167	31.5	0.15	156.2
41	8256	<1	0.46	27	0.2	<0.02	1.3
42	8257	<1	0.11	<5	<0.1	<0.02	0.2
43	8258	<1	0.04	<5	0.1	0.02	<0.1
44	8259	<1	<0.02	<5	0.2	<0.02	<1
45	8260	<1	0.04	8	<0.1	<0.02	0.1
46	8261	<1	0.03	<5	0.3	<0.02	<0.1
47	8262	<1	0.04	9	0.3	0.4	<0.1
48	8263	<1	0.04	8	0.3	<0.02	0.1
49	8264	1	0.03	15	1.1	0.02	0.2
50	8265	<1	0.03	5	0.2	<0.02	0.1
[illegible]1	8266	1	0.02	100	4.6	0.04	12.2
[illegible]	8267	<1	0.03	58	5.3	0.08	14.8

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
53	8268	1	0.03	34	3.0	0.1	5.3
54	8269	<1	0.03	49	2.6	0.08	8.2
55	8270	1	0.03	53	6.2	0.07	24.3
56	8271	<1	0.03	24	1.9	0.03	3.4
57	8272	<1	0.03	33	0.6	0.03	1.4
58	8273	<1	0.04	64	4.3	0.07	9.7
59	8274	<1	0.03	24	1.3	0.07	2.5
60	8275	<1	0.03	13	2.2	0.04	7.7
61	8276	<1	0.04	47	4.9	0.02	10.5
62	8277	<1	0.03	74	3.3	0.04	23.3
63	8278	<1	0.17	189	7.1	0.03	25.2
64	8279	<1	0.15	129	11.0	0.08	18.4
65	8280	<1	0.05	27	2.1	0.05	6.6
66	8281	<1	0.06	7	0.3	<0.02	2.8
67	8282	<1	0.05	5	0.2	<0.02	1.7
68	8283	<1	0.07	11	0.3	0.03	0.3
69	8284	<1	0.1	8	0.2	<0.02	0.2
70	8285	<1	<0.02	12	0.1	<0.02	<0.1
71	8286	<1	0.05	23	0.3	0.02	2.0
72	8287	<1	0.03	23	0.4	<0.02	1.8
73	8288	<1	0.04	23	0.3	<0.02	1.4
74	8289	<1	0.03	40	0.8	<0.02	1.8
	8290	<1	0.04	11	0.3	<0.02	1.7
76	8291	<1	0.05	27	0.6	<0.02	1.9
77	8292	<1	0.04	10	0.3	<0.02	1.9
78	8293	<1	0.03	24	0.4	0.02	1.7
79	8294	<1	0.04	40	1.0	0.04	2.3
80	8295	<1	0.05	126	2.9	0.02	3.5
81	8296	<1	0.04	92	1.1	<0.02	3.7
82	8297	<1	0.04	114	6.8	0.11	24.4
83	8298	<1	0.05	11	0.3	<0.02	1.0
84	8299	<1	0.05	27	0.7	0.03	2.8
85	8300	<1	0.04	12	1.4	0.03	4.3
86	8301	<1	0.04	16	4.6	0.05	10.2
87	8302	<1	0.05	13	1.3	<0.02	4.3
88	8303	<1	0.06	13	0.8	0.05	1.9
89	8304	<1	0.05	19	0.7	0.02	1.5

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
QC DATA:							
Resplit:							
36	8251	<1	0.15	41	2.3	0.04	0.2
71	8286	<1	0.05	23	0.3	<0.02	1.9
Repeat:							
50	8265	<1	0.04	7	0.2	<0.02	0.1
80	8295	<1	0.05	118	2.9	<0.02	3.5
Standard:							
Standard		1	1.09	216	1.3	1.01	6.6
Standard		<1	1.15	222	1.3	0.96	6.5

NOTE: Results run by ACME Analytical Laboratories Ltd.

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E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

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Phone (250) 573-5700 Fax (250) 573-4557
email: ecotech@direct.ca

CERTIFICATE OF ANALYSIS AK 2002-124B

24-Jul-02

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

ATTENTION: Louis Doyle

No. of samples received: 5
Sample type: Core
Project #: Frank Creek
Shipment #: None Given
Samples submitted by: Chris Wild

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
61	08235	<1	0.03	10	2.50	0.03	0.3
62	08236	<1	0.05	26	1.10	0.05	5
63	08237	<1	0.04	70	0.90	0.02	1.4
64	08238	<1	0.06	116	1.20	0.05	2.2
65	08239	<1	0.04	80	1.00	0.03	1.0

NOTE: Results run by ACME Analytical Laboratories Ltd.

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CERTIFICATE OF ANALYSIS AK 2002-124

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

24-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 60
Sample type: Core
Project #: ACE
Shipment #: None Given
Samples submitted by: Chris Wild

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
1	08151 Ace 02-01 13.0-14.5m	<1	0.50	<5	0.3	0.07	0.3
2	08152 Ace 02-01 17.37-19.0m	<1	0.32	<5	0.3	<0.02	0.2
3	08153 Ace 02-01 29.0-30.5m	<1	0.06	<5	1.10	<0.02	0.2
4	08154 Ace 02-01 37.5-39.0m	<1	0.06	<5	5.10	0.03	0.1
5	08155 Ace 02-01 45.0-46.5m	<1	0.07	<5	0.60	<0.02	0.1
6	08156 Ace 02-01 47.5-49.0m	<1	0.06	<5	6.00	0.03	0.1
7	08157 Ace 02-01 49.0-49.7m	<1	0.07	<5	4.70	0.08	0.2
8	08158	<1	0.02	<5	1.10	0.09	<0.1
9	08159 Ace 02-01 50.7-52.1m	1	0.05	<5	5.20	0.05	0.2
10	08160 Ace 02-01 52.1-54.0m	<1	0.03	<5	0.40	0.11	0.1
11	08161 Ace 02-01 54.0-55.2m	<1	0.09	<5	1.90	0.06	0.2
12	08162 Ace 02-01 55.2-57.0m	<1	0.10	<5	4.20	0.05	0.2
13	08163 Ace 02-01 57.0-58.8m	1	0.09	<5	3.80	0.06	0.3
14	08164 Ace 02-01 58.8-61.0m	<1	0.12	<5	3.80	0.05	0.2
15	08165 Ace 02-01 61.0-63.0m	1	0.24	<5	3.40	0.04	0.4
16	08166 Ace 02-01 63.0-65.3m	1	0.27	<5	3.40	0.06	0.3
17	08167 Ace 02-01 65.3-67.5m	2	0.31	<5	8.90	0.12	0.2
18	08168 Ace 02-01 76.0-77.5m	<1	0.35	<5	<1	<0.02	0.4
19	08169 Ace 02-01 88.0-89.5m	<1	0.07	<5	0.20	<0.02	0.2
20	08170 Ace 02-01 103.6-105.1m	<1	0.12	<5	0.70	0.02	0.2
21	08171 Ace 02-01 130.0-131.5m	2	0.08	<5	0.40	0.03	0.2
22	08172 Ace 02-01 139.5-141.0m	<1	0.05	<5	0.10	<.02	0.2
23	08173	<1	0.43	<5	0.40	0.03	0.5
24	08174	8	0.33	<5	1.60	0.76	0.6
25	08175	1	0.13	<5	0.60	0.04	0.3
26	08176	1	0.06	<5	0.30	<.02	0.2
27	08177	<1	0.04	<5	0.10	<.02	0.2
28	08178	2	0.05	<5	0.50	0.06	0.2
29	08179	<1	0.05	<5	<0.1	<0.02	0.2

BARKER MINERALS LTD. AK 2002-124 24-Jul-02

ET #.	Tag #	B (ppm)	Tl (ppm)	Hg (ppb)	Se (ppm)	Te (ppm)	Sn (ppm)
30	08180	<1	0.08	<5	15.20	0.34	0.7
31	08181	<1	0.13	<5	10.60	0.27	1.2
32	08182	1	0.91	<5	9.10	0.15	1.3
33	08183	1	0.07	<5	3.20	0.04	0.2
34	08184	1	0.22	6	5.60	0.13	0.3
35	08185	1	0.15	<5	3.70	0.07	0.3
36	08186	2	0.16	5	3.90	0.06	0.3
37	08187	<1	0.05	<5	2.20	0.05	0.5
38	08188	<1	0.04	<5	2.00	0.08	0.3
39	08189	1	0.37	<5	3.40	0.04	0.5
40	08190	<1	0.10	<5	0.20	0.02	0.2
41	08191	<1	0.08	<5	2.70	0.07	0.2
42	08192	<1	0.31	<5	2.60	0.06	0.3
43	08193	<1	0.9	<5	2.90	0.06	0.6
44	08194	<1	0.98	<5	0.50	0.02	0.3
45	08195	<1	0.48	<5	2.40	0.06	0.4
46	08196	<1	0.05	<5	0.40	<0.02	0.2
47	08197	<1	0.04	<5	0.30	<0.02	0.1
48	08198	<1	0.03	<5	0.20	0.02	0.1
49	08212	<1	0.04	<5	6.60	0.06	0.2
50	08213	1	0.03	<5	5.60	0.10	0.2
51	08214	<1	0.14	<5	27.60	0.97	0.3
52	08215	<1	0.20	<5	4.90	0.24	0.2
53	08216	<1	0.12	<5	1.20	0.06	0.4
54	08217	<1	0.07	<5	7.00	0.10	0.3
55	08229	<1	0.04	<5	0.30	0.02	0.1
56	08230	<1	0.12	<5	4.00	0.07	0.5
57	08231	<1	0.23	<5	4.20	0.10	0.5
58	08232	<1	0.03	<5	0.30	<0.02	0.2
59	08233	<1	0.08	<5	2.60	0.04	0.2
60	08234	<1	0.21	<5	2.50	0.04	0.4
QC DATA:							
Resplit:							
1	08151 Ace 02-01 13.0-14.5m	<1	0.50	<5	0.2	0.08	0.4
36	08186	2	0.15	<5	3.70	0.07	0.4
Repeat:							
17	08167 Ace 02-01 65.3-67.5m	<1	0.3	<5	8.70	0.12	0.2
50	08213	1	0.03	<5	5.90	0.08	0.1
Standard:							
Standard		3	1.27	236	1.40	0.98	7.0
Standard		2	1.23	234	1.40	0.99	6.9

NOTE: Results run by ACME Analytical Laboratories Ltd.

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CERTIFICATE OF ANALYSIS AK 2002-131

BARKER MINERALS LTD.
22117 37A Ave
Langley, BC
V2Z 1N9

25-Jul-02

ATTENTION: Louis Doyle

No. of samples received: 65
Sample Type: Core
Project #: Frank Creek
Drill Hole #: FC-02-01 to FC-02-06
Samples submitted by: Chris Wild

Note: Values expressed in percent

ET #.	Tag #	BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
27	8242	0.14	0.22	52.83	0.33	7.34	4.36	13.17	5.22	0.52	0.29	3.76	11.84
31	8246	0.07	0.06	76.61	0.19	3.95	1.39	8.81	1.85	0.46	0.33	1.24	5.03
32	8247	0.08	0.31	31.44	0.17	9.81	7.20	9.25	15.39	2.06	1.00	0.19	23.11
35	8250	0.21	0.28	25.69	0.15	9.60	8.00	8.87	17.91	1.89	0.70	0.56	26.14
37	8252	0.07	0.06	79.53	0.07	2.46	1.56	6.59	2.77	0.33	1.32	1.18	4.05
38	8253	0.05	0.22	34.70	0.18	9.34	6.41	10.57	12.71	1.20	1.70	1.10	21.82
39	8254	0.06	0.07	79.21	0.04	3.42	1.08	9.28	0.27	0.43	0.53	2.18	3.41
41	8256	0.15	0.55	46.74	0.12	8.54	6.98	12.98	6.96	0.77	0.09	1.11	15.01
43	8258	0.06	0.12	34.11	0.26	10.71	10.22	7.52	12.25	1.52	0.19	0.79	22.27
44	8259	0.03	0.20	40.65	0.21	10.92	7.37	10.50	10.26	1.96	1.37	0.22	16.31
45	8260	0.09	0.14	70.02	0.06	5.40	1.94	11.80	0.80	0.63	1.76	2.17	5.20
47	8262	0.06	0.07	33.23	0.35	8.46	8.87	4.86	17.12	0.83	0.10	0.83	25.23
50	8265	0.09	0.16	53.35	0.14	7.04	2.71	9.47	9.62	0.98	1.13	1.98	13.31
56	8271	0.05	0.03	69.43	0.16	11.84	2.50	5.99	0.12	0.29	0.43	0.75	8.41
59	8274	0.05	0.04	63.30	0.23	14.19	3.59	7.42	0.25	0.36	0.43	0.73	9.42
66	8281	0.09	0.02	66.39	0.17	8.98	2.84	11.47	0.28	0.52	0.32	2.30	6.60
67	8282	0.11	0.04	67.28	0.22	5.15	2.57	9.41	4.42	0.44	0.41	2.24	7.71
68	8283	0.13	0.06	55.63	0.13	8.03	3.05	17.07	1.95	0.80	0.61	4.43	8.12
69	8284	0.10	0.12	64.49	0.11	6.62	2.77	15.62	0.25	0.65	0.24	3.84	5.20
70	8285	0.02	0.21	40.09	0.19	8.88	6.61	12.18	9.47	1.58	1.24	0.45	19.08
71	8286	0.10	0.07	62.08	0.17	9.17	3.22	12.43	0.20	0.55	0.49	3.12	8.40
74	8289	0.08	0.15	61.37	0.17	11.08	3.36	12.99	0.39	0.75	0.30	2.63	6.72
77	8292	0.06	0.08	62.55	0.18	10.49	3.39	13.23	0.20	0.55	0.24	2.64	6.41
79	8294	0.08	0.09	67.49	0.14	8.46	2.95	12.40	0.30	0.51	0.20	2.46	4.91
83	8298	0.11	0.20	43.23	0.43	10.05	7.85	14.77	5.45	1.39	0.46	2.68	13.39
84	8299	0.12	0.05	57.78	0.18	9.79	3.22	14.54	0.33	0.60	0.69	3.62	9.06
89	8304	0.10	0.07	59.81	0.16	8.36	2.29	13.40	3.23	0.53	0.49	3.53	8.02

QC DATA:
Standard:

		BaO	P205	SiO2	MnO	Fe203	MgO	Al203	CaO	TiO2	Na2O	K20	L.O.I.
Mrg-1		0.01	0.03	38.49	0.18	17.73	13.42	8.47	14.79	3.74	0.77	0.14	2.22
Sy-4		0.03	0.14	49.38	0.12	6.56	0.66	20.98	8.33	0.31	7.25	1.70	4.56

ECO TECH LABORATORY LTD.
Jutta Jealouse
B.C. Certified Assayer

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E-mail - Barker Minerals
CC: Wildrock Resources - Email & Mail

ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project :
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 25-JUL-2002
Invoice No. : I0220690
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0220690

SAMPLE	PREP CODE		Ag g/t	Cu %	Pb %	Zn %
02104	212	--	115	3.10	-----	2.03
02105	212	--	134	4.31	-----	1.87
02107	212	--	>1500	-----	>50.0	-----

CERTIFICATION:

OVERLIMITS from A0219735

ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project :
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 25-JUL-2002
Invoice No. : I0219735
P.O. Number :
Account :PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0219735

SAMPLE	PREP CODE	Weight Kg	Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %	Ag ppm	Ba ppm	Ce ppm
02101	413 402	1.00	0.73	0.01	0.24	< 0.01	29.68	0.10	0.90	0.07	< 0.01	< 0.01	49.91	< 0.01	0.02	16.49	98.15	6	141.5	6.5
02102	413 402	1.28	0.70	0.02	0.33	< 0.01	37.91	0.11	0.97	0.11	0.01	0.01	37.51	< 0.01	0.02	20.85	98.55	3	121.0	7.5
02103	413 402	0.84	0.37	< 0.01	0.14	0.01	29.77	0.05	0.61	0.06	< 0.01	< 0.01	51.61	< 0.01	< 0.01	16.72	99.34	3	30.0	6.0
02104	413 402	2.40	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	36	18.0	3.5
02105	413 402	2.52	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	28	51.0	4.0
02106	413 402	0.54	0.24	< 0.01	0.01	0.02	15.27	0.03	< 0.01	0.03	0.07	< 0.01	80.55	< 0.01	0.01	3.08	99.31	1	17.5	2.5
02107	402 400	0.10	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----
02108	402 400	0.10	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----
02109	413 402	0.40	13.69	0.10	1.46	0.01	7.38	3.27	1.68	0.09	0.56	0.12	65.95	0.01	0.82	3.69	98.83	< 1	700	79.5
02110	413 402	0.66	0.41	< 0.01	0.01	0.03	14.26	0.08	< 0.01	0.03	0.05	< 0.01	76.43	< 0.01	< 0.01	7.86	99.16	1	29.5	3.0
A0203-22A	413 402	2.80	17.52	0.33	2.40	0.01	7.06	4.41	1.75	0.11	2.46	0.07	57.62	0.02	0.70	5.12	99.58	< 1	3350	107.5
A0203-23	413 402	1.80	11.15	0.37	3.23	0.03	4.91	1.74	1.14	0.13	3.38	0.36	68.89	0.03	0.40	3.62	99.38	< 1	3260	42.0
A0203-50	413 402	2.52	19.37	0.12	0.61	0.01	8.97	3.92	2.59	0.07	1.43	0.11	56.53	0.02	1.17	3.93	98.85	< 1	994	112.5
A0203-51	413 402	2.42	18.94	0.08	0.69	0.01	9.32	3.25	2.52	0.11	2.08	0.12	56.91	0.01	1.23	3.81	99.08	< 1	710	105.5
A0203-52	413 402	1.50	19.37	0.09	0.80	0.01	9.40	3.50	2.59	0.10	1.65	0.12	56.28	0.02	1.26	4.08	99.27	< 1	751	119.0
A0203-53	413 402	0.52	20.26	0.14	3.05	0.01	10.61	3.75	2.88	0.10	3.53	0.17	48.38	0.04	1.03	5.17	99.12	< 1	1165	84.0
A0203-54	413 402	2.26	20.59	0.15	0.28	0.01	8.30	4.92	2.22	0.06	1.30	0.12	57.11	0.01	1.07	3.49	99.63	< 1	1260	107.0
A0203-55	413 402	1.18	19.64	0.13	0.30	0.02	7.95	5.42	2.30	0.07	0.43	0.15	58.82	0.02	0.91	3.56	99.72	< 1	1125	108.0
A0203-56	413 402	1.06	16.71	0.07	0.62	0.01	9.39	1.18	2.63	0.07	7.64	0.11	56.97	0.03	1.04	3.03	99.50	< 1	532	95.5
A0203-57	413 402	1.68	19.20	0.16	0.83	0.01	7.74	3.38	2.33	0.07	6.09	0.13	54.84	0.03	0.93	2.88	98.62	< 1	1420	117.5
A0203-58	413 402	1.94	15.67	0.09	2.96	0.01	6.82	1.53	2.29	0.10	7.09	0.11	57.65	0.05	0.74	4.19	99.30	< 1	744	82.0
A0203-59	413 402	1.96	19.25	0.08	0.92	0.02	9.15	3.16	2.52	0.13	1.99	0.12	56.95	0.01	1.29	3.59	99.18	< 1	792	116.0
A0203-60	413 402	1.06	17.70	0.09	0.65	0.01	8.16	3.04	2.64	0.09	1.89	0.10	60.49	0.01	1.11	3.39	99.37	< 1	727	87.0
A0203-61	413 402	2.04	16.61	0.08	1.03	0.03	8.92	2.73	2.37	0.15	1.48	0.11	61.13	0.01	1.21	3.49	99.35	< 1	623	92.0
A0203-62	413 402	2.02	19.65	0.09	0.78	0.02	9.81	3.37	2.67	0.16	1.76	0.12	56.06	0.01	1.31	3.72	99.53	< 1	730	113.5
A0203-63	413 402	2.52	19.31	0.10	0.52	0.02	9.58	3.23	2.61	0.13	2.11	0.14	56.56	0.01	1.32	3.41	99.05	< 1	785	132.5
A0203-64	413 402	1.38	18.46	0.12	0.60	0.01	8.70	3.62	2.65	0.09	1.28	0.10	58.70	0.01	1.08	3.81	99.23	< 1	996	111.0
A0203-65	413 402	0.68	19.26	0.08	0.73	0.01	10.04	2.72	2.81	0.13	2.69	0.13	56.11	0.02	1.69	3.47	99.89	< 1	565	100.0
A0203-66	413 402	0.60	18.80	0.09	0.69	0.01	9.68	3.36	2.52	0.14	1.50	0.11	56.86	0.01	1.36	3.75	98.88	< 1	637	111.5

CERTIFICATION: [signature]

* UNABLE TO FUSE SAMPLES 02104 AND 02105 FOR WHOLE ROCK.

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SAMPLE	PREP CODE	Co ppm	Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm	Ni ppm	Pb ppm	Pr ppm
02101	9413 9402	284	120	0.9	5830	2.0	0.4	0.2	2	0.9	< 1	0.1	10.0	< 0.1	20	1	3.5	120	375	1.1
02102	9413 9402	361	80	0.9	1825	1.1	0.5	0.1	2	0.9	< 1	0.1	11.0	< 0.1	16	1	4.0	155	295	1.1
02103	9413 9402	301	220	0.4	5430	0.3	0.1	< 0.1	1	0.4	< 1	< 0.1	7.0	< 0.1	16	< 1	1.5	165	235	0.4
02104	9413 9402	89.0	20	0.2	>10000	0.6	0.1	< 0.1	1	0.4	< 1	< 0.1	6.0	< 0.1	14	< 1	1.5	20	3770	0.5
02105	9413 9402	96.0	30	0.3	>10000	1.5	0.3	0.1	1	0.6	< 1	0.1	6.5	< 0.1	16	< 1	2.0	20	2860	0.6
02106	9413 9402	35.5	220	0.2	545	0.3	< 0.1	< 0.1	1	0.1	< 1	< 0.1	7.0	< 0.1	12	< 1	1.0	50	80	0.4
02107	9402 9400	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----
02108	9402 9400	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----
02109	9413 9402	20.0	180	4.3	70	5.7	3.5	1.2	21	7.1	3	1.1	44.0	0.5	6	14	37.5	40	90	9.7
02110	9413 9402	107.5	310	0.3	225	0.1	< 0.1	< 0.1	1	0.1	< 1	< 0.1	6.0	< 0.1	12	< 1	1.0	125	345	0.4
A0203-22A	9413 9402	22.0	130	4.8	100	4.6	2.6	1.3	25	7.7	3	0.9	59.0	0.4	16	16	44.5	50	25	13.4
A0203-23	9413 9402	12.0	180	2.2	60	3.5	2.3	0.9	16	4.1	1	0.7	27.0	0.3	106	9	19.5	70	90	5.2
A0203-50	9413 9402	25.0	130	4.0	110	6.5	3.7	1.8	27	9.2	5	1.2	58.0	0.5	6	19	50.0	60	30	12.9
A0203-51	9413 9402	27.0	140	3.2	50	6.5	3.8	1.7	27	9.0	5	1.2	55.0	0.5	8	21	49.0	85	45	12.7
A0203-52	9413 9402	26.0	120	3.5	55	7.1	3.9	1.9	28	10.2	5	1.3	63.0	0.5	6	21	54.5	60	20	13.9
A0203-53	9413 9402	28.5	110	5.2	120	5.0	3.2	1.3	28	6.7	5	1.0	44.5	0.7	22	20	36.5	55	65	9.7
A0203-54	9413 9402	21.0	130	5.4	35	6.1	3.5	1.6	26	8.3	5	1.2	51.5	0.5	6	19	44.5	45	15	11.7
A0203-55	9413 9402	24.5	160	5.3	40	6.5	4.1	1.7	29	9.5	4	1.3	62.0	0.5	6	22	49.5	55	45	13.8
A0203-56	9413 9402	26.5	120	6.7	80	6.5	3.7	1.6	23	8.2	4	1.3	48.5	0.4	6	23	40.5	55	35	11.5
A0203-57	9413 9402	23.5	120	13.6	75	5.6	3.0	1.7	30	9.4	4	1.0	61.0	0.4	6	23	51.0	50	35	13.8
A0203-58	9413 9402	18.0	120	8.5	80	6.4	3.7	1.4	21	7.5	3	1.2	45.0	0.5	8	14	37.0	50	25	10.0
A0203-59	9413 9402	26.0	140	3.3	95	7.4	4.0	2.1	27	9.8	5	1.4	59.5	0.5	6	22	50.0	50	20	13.8
A0203-60	9413 9402	22.0	120	3.2	85	5.9	3.3	1.5	24	7.5	5	1.1	45.0	0.4	6	18	38.0	50	15	10.2
A0203-61	9413 9402	23.0	150	2.9	35	6.3	3.4	1.6	23	8.0	4	1.2	46.5	0.5	6	19	40.5	45	25	10.7
A0203-62	9413 9402	28.5	120	3.5	40	7.5	4.0	1.9	27	9.4	5	1.4	57.5	0.5	8	23	49.5	55	15	13.0
A0203-63	9413 9402	28.5	120	3.3	65	8.0	4.4	2.4	28	11.6	5	1.5	71.0	0.5	6	25	60.0	60	15	16.7
A0203-64	9413 9402	25.0	120	3.8	120	6.1	3.5	1.7	27	8.6	4	1.1	57.0	0.4	6	20	48.5	55	10	12.7
A0203-65	9413 9402	28.5	130	2.9	75	7.2	4.1	1.9	27	9.1	5	1.3	51.0	0.6	6	22	45.0	60	10	11.7
A0203-66	9413 9402	26.5	120	3.7	40	7.3	4.2	1.9	28	9.7	5	1.3	61.5	0.5	6	24	51.5	60	10	13.8

CERTIFICATION:

* UNABLE TO FUSE SAMPLES 02104 AND 02105 FOR WHOLE ROCK.

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SAMPLE	PREP CODE	Rb ppm	Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm	Au CN g/t	Au ppb ICP	Pt ppb ICP
02101	9413 9402	4.8	0.6	16	16.7	0.5	0.1	< 1	< 0.5	< 0.1	4.5	15	7	5.0	0.3	455	27.0	-----	48	< 10
02102	9413 9402	4.2	0.6	8	7.1	1.0	0.1	< 1	< 0.5	< 0.1	3.0	40	9	5.0	0.3	440	14.5	-----	36	< 10
02103	9413 9402	1.0	0.3	9	2.6	< 0.5	< 0.1	< 1	< 0.5	< 0.1	2.0	10	8	3.0	0.1	295	6.0	-----	28	< 10
02104	9413 9402	1.6	0.3	115	7.2	< 0.5	< 0.1	< 1	3.5	< 0.1	< 0.5	< 5	3	1.5	0.1	>10000	3.0	-----	748	< 10
02105	9413 9402	4.0	0.4	192	5.0	< 0.5	< 0.1	< 1	4.5	< 0.1	1.5	< 5	4	3.0	0.1	>10000	4.0	-----	704	< 10
02106	9413 9402	1.6	0.1	4	4.1	< 0.5	< 0.1	< 1	< 0.5	< 0.1	< 0.5	15	4	0.5	< 0.1	110	5.0	-----	>10000	10
02107	9402 9400	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	0.04	144	180
02108	9402 9400	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	-----	2.81	3740	< 60
02109	9413 9402	136.0	7.1	4	148.0	1.0	0.9	12	< 0.5	0.4	2.5	100	7	30.0	2.9	85	124.0	-----	1880	< 5
02110	9413 9402	3.6	0.1	1	5.1	< 0.5	< 0.1	< 1	< 0.5	< 0.1	< 0.5	< 5	4	0.5	0.1	30	6.5	-----	62	20
A0203-22A	9413 9402	207	7.9	14	186.0	1.0	0.9	17	0.5	0.3	6.0	275	19	23.5	2.5	75	82.5	-----	34	< 5
A0203-23	9413 9402	94.2	3.6	9	220	0.5	0.5	7	< 0.5	0.3	8.0	110	10	22.0	1.9	220	51.5	-----	2	10
A0203-50	9413 9402	165.5	9.0	3	137.0	2.0	1.2	16	0.5	0.4	3.0	140	4	32.5	3.0	120	159.5	-----	< 2	45
A0203-51	9413 9402	134.5	8.9	3	150.5	1.5	1.1	16	< 0.5	0.4	3.0	135	4	33.5	3.1	145	157.5	-----	< 2	< 5
A0203-52	9413 9402	149.5	9.6	3	159.0	1.5	1.2	17	< 0.5	0.5	3.0	125	3	34.5	3.2	145	157.5	-----	< 2	10
A0203-53	9413 9402	160.0	6.4	4	287	1.5	0.8	13	< 0.5	0.5	3.5	355	4	26.5	3.4	170	145.0	-----	< 2	5
A0203-54	9413 9402	189.0	8.0	3	114.5	1.5	1.1	17	0.5	0.4	3.0	105	3	30.5	2.9	190	142.5	-----	< 2	< 5
A0203-55	9413 9402	196.0	9.4	4	98.8	1.5	1.2	16	0.5	0.5	3.0	120	2	33.0	3.5	145	152.5	-----	10	20
A0203-56	9413 9402	78.2	7.9	3	245	1.5	1.1	12	< 0.5	0.4	2.5	255	4	30.0	3.0	215	157.0	-----	< 2	< 5
A0203-57	9413 9402	164.5	9.3	4	208	1.5	1.1	17	0.5	0.4	3.5	265	8	26.0	2.7	140	147.5	-----	< 2	< 5
A0203-58	9413 9402	84.6	7.2	3	369	1.5	1.0	12	< 0.5	0.4	3.5	235	5	31.0	3.3	255	106.0	-----	< 2	< 5
A0203-59	9413 9402	139.0	9.3	3	151.5	1.5	1.3	17	< 0.5	0.5	3.0	125	3	34.0	3.4	125	201	-----	< 2	< 5
A0203-60	9413 9402	126.5	7.1	3	118.5	1.5	1.0	14	< 0.5	0.4	2.5	115	3	29.0	2.8	120	174.0	-----	< 2	< 5
A0203-61	9413 9402	119.0	7.6	3	129.0	1.5	1.1	14	< 0.5	0.4	2.5	110	2	30.0	3.1	120	138.0	-----	< 2	< 5
A0203-62	9413 9402	145.5	8.8	3	139.0	2.0	1.2	17	< 0.5	0.5	3.0	135	3	35.5	3.2	135	153.5	-----	< 2	5
A0203-63	9413 9402	136.0	11.3	3	134.0	1.5	1.5	19	< 0.5	0.5	3.0	140	3	36.5	3.7	140	179.0	-----	< 2	< 5
A0203-64	9413 9402	156.0	8.5	3	122.5	1.5	1.1	18	< 0.5	0.4	3.0	135	3	30.5	2.8	125	149.0	-----	< 2	10
A0203-65	9413 9402	117.0	8.5	3	140.0	1.5	1.2	14	< 0.5	0.5	2.5	160	2	35.0	3.5	125	178.5	-----	< 2	< 5
A0203-66	9413 9402	143.0	9.8	3	132.5	2.0	1.2	16	0.5	0.5	3.0	145	3	36.0	3.4	135	168.5	-----	< 2	< 5

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SAMPLE	PREP CODE	Pd ppb ICP	Au FA g/t	Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)	Ge ppm (ICP)	Hf ppm
02101	9413 9402	20	-----	10.75	0.33	3.0	64.5	0.15	35.2	0.17	1.78	3.69	302.0	104	0.60	9210	21.7	1.05	0.60	0.3
02102	9413 9402	16	-----	5.34	0.29	2.8	36.5	0.05	36.4	0.21	1.56	4.15	364.0	61	0.75	2680	>25.0	1.05	0.75	0.2
02103	9413 9402	12	-----	5.72	0.11	1.8	23.0	0.05	29.0	0.09	1.24	2.37	330.0	159	0.45	6920	21.6	0.45	0.60	< 0.1
02104	9413 9402	4	-----	>100.0	0.07	806	8.5	< 0.05	830	0.01	58.9	0.51	166.7	58	0.15	>10000	>25.0	1.20	1.20	< 0.1
02105	9413 9402	< 4	-----	>100.0	0.17	721	13.5	< 0.05	646	0.03	56.5	1.93	182.6	88	0.20	>10000	>25.0	1.55	1.20	< 0.1
02106	9413 9402	< 4	13.80	5.84	0.03	3.2	0.5	< 0.05	438	< 0.01	0.36	0.11	31.3	241	0.15	712.0	11.95	0.25	0.25	< 0.1
02107	9402 9400	12	-----	>100.0	0.14	219	47.0	< 0.05	7280	0.06	179.0	15.65	11.1	55	0.10	529.0	2.51	0.50	4.55	< 0.1
02108	9402 9400	12	-----	3.90	0.40	9.8	16.5	< 0.05	14.25	0.09	0.40	12.00	54.6	6280	0.20	30.4	>25.0	14.05	1.50	0.3
02109	9413 9402	< 2	-----	1.32	7.08	< 0.2	660.6	2.45	80.7	1.10	0.42	75.4	20.0	123	4.40	70.8	5.43	20.25	0.25	0.6
02110	9413 9402	20	-----	1.90	0.15	3.4	23.5	0.05	7.75	< 0.01	0.32	0.40	113.8	192	0.20	336.0	11.65	0.60	0.45	< 0.1
A0203-22A	9413 9402	< 2	-----	0.26	9.32	0.2	732.0	5.10	4.15	1.80	0.42	99.0	19.9	88	5.25	111.1	5.02	27.10	0.25	0.6
A0203-23	9413 9402	6	-----	0.48	5.89	< 0.2	182.5	5.00	6.34	2.30	4.46	36.5	11.5	126	2.15	56.2	3.45	15.30	0.20	0.4
A0203-50	9413 9402	< 2	-----	0.34	10.60	< 0.2	1093.5	3.05	0.36	0.39	0.04	103.5	25.8	112	4.25	115.6	6.61	29.10	0.30	0.8
A0203-51	9413 9402	< 2	-----	0.32	10.45	< 0.2	794.8	3.15	0.45	0.53	0.06	120.5	28.6	119	3.60	39.8	6.92	31.05	0.30	0.6
A0203-52	9413 9402	< 2	-----	0.26	10.60	< 0.2	810.4	2.90	0.22	0.63	0.06	118.5	27.1	99	3.75	44.8	6.86	29.90	0.35	0.5
A0203-53	9413 9402	< 2	-----	0.62	10.55	< 0.2	619.3	3.75	1.45	2.30	0.40	85.6	28.8	95	5.45	132.1	7.35	30.65	0.30	0.4
A0203-54	9413 9402	< 2	-----	0.50	11.70	< 0.2	1441.0	3.60	0.30	0.23	0.06	118.0	22.7	117	6.10	40.4	6.36	31.55	0.30	0.6
A0203-55	9413 9402	6	-----	0.68	11.25	< 0.2	1206.5	3.65	0.55	0.24	0.02	125.0	25.2	113	5.90	42.2	6.16	31.60	0.35	0.8
A0203-56	9413 9402	< 2	-----	0.34	8.68	0.6	252.0	2.30	1.32	0.46	0.48	99.5	25.6	95	7.20	69.7	6.66	24.80	0.30	0.6
A0203-57	9413 9402	< 2	-----	0.28	10.70	< 0.2	352.5	3.70	0.49	0.68	0.14	116.5	23.8	101	15.25	58.9	5.91	33.70	0.30	0.8
A0203-58	9413 9402	2	-----	0.28	8.16	0.6	412.5	2.45	0.78	2.20	0.88	79.7	18.3	100	8.70	83.9	4.73	21.30	0.25	0.7
A0203-59	9413 9402	< 2	-----	0.36	10.00	< 0.2	798.4	2.70	0.15	0.66	0.08	117.0	25.7	121	3.55	57.7	6.43	29.00	0.30	0.6
A0203-60	9413 9402	< 2	-----	0.24	9.42	< 0.2	735.2	2.40	0.11	0.41	0.04	90.4	21.6	99	3.40	61.6	5.74	25.25	0.25	0.6
A0203-61	9413 9402	< 2	-----	0.26	9.14	< 0.2	670.7	2.80	0.19	0.78	0.08	93.5	24.3	126	3.25	33.2	6.53	26.20	0.30	0.5
A0203-62	9413 9402	< 2	-----	0.36	11.05	< 0.2	800.4	3.10	0.11	0.58	0.06	119.5	28.1	109	3.95	37.2	7.28	31.35	0.35	0.6
A0203-63	9413 9402	< 2	-----	0.32	10.90	< 0.2	838.4	2.70	0.11	0.41	0.04	141.5	25.7	91	3.65	54.7	7.08	31.20	0.35	0.6
A0203-64	9413 9402	< 2	-----	0.26	10.45	< 0.2	1084.0	3.20	0.18	0.45	0.02	107.0	26.1	97	4.10	89.4	6.34	29.10	0.30	0.7
A0203-65	9413 9402	< 2	-----	0.28	10.60	< 0.2	622.6	2.65	0.15	0.56	0.02	110.0	29.9	112	3.30	75.2	7.36	28.80	0.30	0.7
A0203-66	9413 9402	< 2	-----	0.26	10.20	< 0.2	673.0	2.65	0.11	0.53	0.02	106.0	26.7	99	3.90	31.2	6.93	29.35	0.30	0.7

CERTIFICATION: Hart Bichler

* UNABLE TO FUSE SAMPLES 02104 AND 02105 FOR WHOLE ROCK.

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project :
Comments: ATTN: LOUIS E. DOYLE

Page Number :1-E
Total Pages :1
Certificate Date: 25-JUL-2002
Invoice No. : I0219735
P.O. Number :
Account : PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0219735

SAMPLE	PREP CODE		In ppm	K % (ICP)	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm	Sn ppm	Sr ppm (ICP)
02101	413	402	0.920	0.07	2.0	2.0	0.55	370	8.90	0.01	0.5	113.0	30	470	2.6	0.002	>10.00	0.25	51	13.4	3.0
02102	413	402	0.575	0.07	2.5	1.6	0.51	695	4.35	0.01	0.6	128.0	60	389	2.4	< 0.002	>10.00	0.25	52	5.2	2.4
02103	413	402	0.825	0.01	1.5	1.0	0.35	300	2.85	< 0.01	0.4	126.5	< 10	239	0.1	< 0.002	>10.00	0.25	52	6.2	1.2
02104	413	402	45.5	0.01	< 0.5	< 0.2	0.05	105	12.30	< 0.01	0.7	25.0	70	6130	0.1	0.002	>10.00	198.95	96	199.5	0.8
02105	413	402	55.6	0.06	0.5	0.2	0.12	175	16.45	< 0.01	0.8	24.8	50	5040	2.5	0.006	>10.00	92.65	112	254	2.8
02106	413	402	0.275	< 0.01	< 0.5	0.2	< 0.01	30	0.85	< 0.01	0.1	35.2	10	105.5	0.1	< 0.002	5.89	1.85	4	2.0	1.0
02107	402	400	0.635	0.03	8.0	0.4	0.01	135	28.80	0.01	0.3	12.8	330	>10000	1.0	< 0.002	>10.00	72.10	>1000	16.0	48.0
02108	402	400	0.035	0.03	6.0	0.8	0.20	720	3.90	0.01	3.0	192.5	120	870	1.3	< 0.002	0.18	0.60	4	2.4	8.8
02109	413	402	0.105	2.87	37.0	17.6	0.91	555	0.55	0.34	9.3	33.4	540	148.5	142.0	< 0.002	1.39	0.40	1	3.4	134.5
02110	413	402	0.025	0.05	< 0.5	1.0	0.01	50	1.35	< 0.01	0.4	120.0	40	526	2.5	< 0.002	9.92	0.80	11	0.4	2.4
A0203-22A	413	402	0.090	3.67	48.0	32.6	1.10	810	11.20	1.77	2.6	48.0	240	19.5	202	0.012	2.06	0.25	5	11.6	180.0
A0203-23	413	402	0.085	1.45	18.5	18.0	0.67	775	93.89	2.46	1.8	61.4	1550	81.0	91.2	0.124	1.58	0.25	3	7.4	217
A0203-50	413	402	0.090	3.30	51.0	31.2	1.56	440	0.85	1.08	10.2	54.6	460	14.5	174.0	0.002	0.99	0.10	1	2.4	148.5
A0203-51	413	402	0.105	2.78	59.0	34.2	1.55	745	0.75	1.46	9.6	55.6	520	28.5	155.0	< 0.002	0.48	0.10	1	2.2	169.0
A0203-52	413	402	0.100	2.95	58.0	36.0	1.57	715	0.60	1.24	8.4	56.7	450	15.0	158.5	< 0.002	0.51	0.10	< 1	2.4	168.5
A0203-53	413	402	0.085	3.01	43.5	39.0	1.63	680	17.10	2.66	9.7	52.4	660	85.5	161.0	0.004	1.70	0.10	4	3.2	295
A0203-54	413	402	0.105	4.24	56.5	36.2	1.37	345	0.45	1.01	19.3	46.4	410	20.0	206	< 0.002	0.46	0.15	1	3.2	133.5
A0203-55	413	402	0.100	4.68	58.5	35.4	1.42	445	0.55	0.31	23.1	50.6	650	58.5	212	< 0.002	0.34	0.20	1	3.2	98.6
A0203-56	413	402	0.070	1.01	44.5	26.4	1.44	410	0.25	5.80	3.3	49.8	450	32.5	78.9	0.002	2.56	0.05	3	1.4	242
A0203-57	413	402	0.110	3.01	55.0	51.4	1.40	425	0.20	4.83	7.4	48.8	590	18.5	177.0	< 0.002	1.64	0.05	2	3.4	228
A0203-58	413	402	0.075	1.27	39.5	22.8	1.26	825	1.80	5.35	3.2	42.8	420	28.5	84.4	0.002	2.29	0.10	3	1.6	346
A0203-59	413	402	0.095	2.65	57.5	30.6	1.43	895	0.45	1.49	12.4	49.4	470	16.5	135.5	< 0.002	0.30	0.05	1	2.2	160.5
A0203-60	413	402	0.080	2.53	46.0	31.2	1.51	530	0.40	1.34	9.0	46.0	420	11.0	126.5	< 0.002	0.24	< 0.05	< 1	2.0	125.5
A0203-61	413	402	0.085	2.43	46.5	30.6	1.46	1120	0.60	1.10	9.4	42.0	540	21.5	125.5	< 0.002	0.19	< 0.05	< 1	2.0	144.5
A0203-62	413	402	0.105	2.93	59.5	35.6	1.69	1225	0.65	1.30	13.5	54.0	670	13.0	153.5	< 0.002	0.24	0.05	1	2.4	167.0
A0203-63	413	402	0.100	2.72	69.0	34.4	1.61	940	0.40	1.62	12.8	49.2	700	12.5	145.5	< 0.002	0.29	< 0.05	1	2.2	143.0
A0203-64	413	402	0.090	3.03	53.5	30.4	1.65	525	0.35	0.96	9.6	53.9	520	11.5	163.5	< 0.002	0.71	0.05	1	2.2	139.0
A0203-65	413	402	0.100	2.30	53.0	34.0	1.69	1015	0.50	2.02	10.8	55.1	650	11.5	127.5	< 0.002	0.27	< 0.05	< 1	2.0	156.0
A0203-66	413	402	0.095	2.80	52.0	32.8	1.51	975	0.55	1.15	9.8	51.7	620	10.0	151.0	< 0.002	0.23	< 0.05	< 1	2.2	135.0

CERTIFICATION:

* UNABLE TO FUSE SAMPLES 02104 AND 02105 FOR WHOLE ROCK.

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project :
Comments: ATTN: LOUIS E. DOYLE

Certificate Date: 25-JUL-2002
Invoice No. : I0219735
P.O. Number :
Account : PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0219735

SAMPLE	PREP CODE		Ta ppm (ICP)	Te ppm (ICP)	Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
02101	9413	9402	< 0.05	1.55	1.2	< 0.01	0.06	4.0	11	3.3	3.5	498	18.0	90
02102	9413	9402	< 0.05	1.70	0.8	< 0.01	0.04	2.8	12	5.3	4.0	450	8.5	50
02103	9413	9402	< 0.05	1.65	< 0.2	< 0.01	< 0.02	2.0	7	3.8	2.5	320	2.0	60
02104	9413	9402	< 0.05	0.75	< 0.2	< 0.01	7.44	0.6	< 1	3.5	0.8	>10000	1.0	16310
02105	9413	9402	< 0.05	0.75	0.2	< 0.01	9.36	2.7	< 1	3.6	3.0	>10000	3.0	18990
02106	9413	9402	< 0.05	47.9	< 0.2	< 0.01	0.20	< 0.1	4	0.2	0.3	126	< 0.5	< 10
02107	9402	9400	< 0.05	386	1.6	0.02	2.12	11.2	20	2.7	5.3	210	2.0	60
02108	9402	9400	0.05	0.65	2.0	0.25	0.08	1.7	2210	3.2	5.5	216	13.0	45400
02109	9413	9402	0.45	2.55	16.0	0.30	0.76	1.9	94	4.4	11.1	72	18.5	240
02110	9413	9402	< 0.05	0.65	0.6	< 0.01	0.12	0.4	10	0.5	0.5	24	0.5	20
A0203-22A	9413	9402	0.15	0.15	21.4	0.14	0.98	5.1	259	11.4	7.7	70	19.5	30
A0203-23	9413	9402	0.05	0.10	5.8	0.07	0.48	7.3	108	5.4	12.6	222	14.5	< 10
A0203-50	9413	9402	0.55	0.05	23.0	0.20	0.86	1.9	140	1.1	17.7	112	24.5	10
A0203-51	9413	9402	0.45	< 0.05	23.8	0.18	0.70	1.7	134	0.4	11.3	138	20.0	< 10
A0203-52	9413	9402	0.40	< 0.05	25.0	0.13	0.78	1.8	130	0.6	8.2	142	19.5	10
A0203-53	9413	9402	0.50	0.25	20.2	0.28	0.78	2.1	350	0.8	12.1	170	12.5	< 10
A0203-54	9413	9402	1.20	< 0.05	26.2	0.62	0.98	2.2	119	1.0	13.1	188	20.5	10
A0203-55	9413	9402	1.40	0.05	25.0	0.57	0.90	2.0	115	0.8	8.4	152	24.0	< 10
A0203-56	9413	9402	0.20	< 0.05	17.4	0.11	0.54	1.3	220	0.5	9.8	212	20.5	10
A0203-57	9413	9402	0.40	< 0.05	23.6	0.30	0.92	2.2	273	1.8	8.8	146	25.0	< 10
A0203-58	9413	9402	0.25	0.05	16.8	0.14	0.52	3.0	219	0.8	11.5	254	22.5	10
A0203-59	9413	9402	0.65	< 0.05	25.6	0.38	0.68	2.1	129	0.4	24.3	112	21.0	10
A0203-60	9413	9402	0.45	< 0.05	21.2	0.23	0.64	1.7	119	0.4	17.6	106	17.0	10
A0203-61	9413	9402	0.45	< 0.05	20.4	0.32	0.62	1.7	119	0.3	21.7	116	18.0	< 10
A0203-62	9413	9402	0.65	< 0.05	25.2	0.42	0.74	2.1	147	0.4	25.3	134	21.5	10
A0203-63	9413	9402	0.65	< 0.05	28.8	0.35	0.70	2.1	141	0.4	26.3	128	22.0	< 10
A0203-64	9413	9402	0.45	< 0.05	23.4	0.20	0.76	1.9	139	0.6	18.0	120	23.0	10
A0203-65	9413	9402	0.55	< 0.05	22.2	0.35	0.58	1.6	167	0.3	14.8	116	23.5	10
A0203-66	9413	9402	0.45	< 0.05	21.8	0.22	0.72	1.7	140	0.4	11.6	122	25.0	10

CERTIFICATION: ________________

* UNABLE TO FUSE SAMPLES 02104 AND 02105 FOR WHOLE ROCK.

ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

.o: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project :
Comments: ATTN: LOUIS E. DOYLE

Pag...ber :1
Total ...ages :1
Certificate Date: 06-AUG-2002
Invoice No. : I0221099
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0221099

SAMPLE	PREP CODE		Ag FA g/t	Pb % con								
02107	212	--	2730	76.51								

CERTIFICATION:

OVERLIMITS from A0220690

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 09-AUG-2002
Invoice No. : I0221520
P.O. Number :
Account : PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %	Ag ppm	Ba ppm	Ce ppm	Co ppm
02111	299	297	1.34	< 0.01	0.95	0.01	23.88	0.06	0.23	0.07	0.10	0.05	59.78	< 0.01	0.04	12.43	98.94	< 1	19.5	14.5	896
02112	299	297	11.04	0.06	0.52	0.01	7.52	1.89	1.89	0.12	1.47	0.10	69.70	< 0.01	0.49	3.30	98.11	< 1	540	67.0	20.0
02113	299	297	2.48	< 0.01	0.20	0.01	6.68	0.05	1.22	0.07	0.40	0.06	84.87	< 0.01	0.11	2.10	98.25	< 1	9.0	43.0	25.5
02114	299	297	4.68	0.04	21.62	0.01	14.14	1.09	9.33	4.23	0.25	0.03	15.11	0.05	0.20	28.84	99.62	1	295	23.5	18.5
02115	299	297	3.16	0.03	3.79	< 0.01	16.96	0.76	2.22	1.87	6.20	0.07	22.87	< 0.01	0.14	16.51	74.58	23	287	17.0	65.5
02116	299	297	3.36	0.02	23.94	< 0.01	17.43	0.74	8.52	5.49	0.48	0.04	7.82	0.05	0.15	31.30	99.34	1	176.0	31.5	15.0
02117	299	297	4.76	0.03	2.83	0.03	7.31	0.44	2.41	0.48	0.09	0.42	73.09	< 0.01	0.24	6.34	98.47	< 1	310	22.5	12.0
02118	299	297	7.99	0.06	1.81	0.02	9.93	0.69	1.82	2.30	1.20	1.26	66.73	0.01	0.39	5.09	99.30	< 1	499	52.0	35.0
02119	299	297	7.68	0.06	1.08	0.03	11.60	0.67	1.96	0.64	0.43	0.73	69.31	< 0.01	0.46	4.98	99.63	< 1	460	60.0	21.5
02120	299	297	1.72	0.03	0.52	< 0.01	50.17	0.32	1.16	0.15	0.06	0.05	16.86	< 0.01	0.15	28.55	99.74	< 1	149.0	9.5	28.5
02121	299	297	5.86	0.09	0.51	0.02	5.58	0.86	1.39	0.13	0.15	0.29	81.17	< 0.01	0.31	3.49	99.85	2	699	27.5	9.0
02122	299	297	13.21	0.18	0.04	< 0.01	19.63	2.12	0.70	0.14	0.28	0.08	53.36	< 0.01	0.57	8.80	99.11	< 1	1455	63.0	29.0
02123	299	297	15.03	0.08	0.27	0.01	2.58	3.23	0.61	0.06	3.57	0.16	70.84	0.01	0.29	1.76	98.50	< 1	824	61.0	3.5
02124	299	297	13.89	0.37	0.34	0.01	2.15	3.81	0.63	0.08	3.09	0.20	72.88	< 0.01	0.26	1.53	99.24	< 1	3110	57.0	4.0
02125	299	297	8.92	0.05	1.71	0.02	10.46	0.78	2.28	0.63	0.97	1.18	67.18	< 0.01	0.61	4.72	99.51	< 1	474	40.5	12.5
TB-1-SCR	299	297	0.66	0.01	0.21	0.03	0.88	0.07	0.06	0.05	1.54	0.08	95.61	< 0.01	0.07	0.28	99.55	< 1	37.5	49.5	0.5
TB-2-SCR	299	297	5.80	0.10	2.16	0.03	5.22	0.89	1.67	0.21	0.26	1.47	76.54	< 0.01	0.25	3.98	98.58	< 1	811	39.0	8.5
TB-3-F6	299	297	15.93	0.27	2.69	0.04	14.77	1.76	7.37	0.22	2.19	0.38	46.96	0.01	2.65	4.34	99.58	< 1	3250	58.0	36.0
TB-4-FC	299	297	3.63	0.02	0.08	0.01	15.63	0.81	1.61	0.17	0.17	0.03	67.89	< 0.01	0.06	9.17	99.28	< 1	202	24.0	48.0
TB-5-FC	299	297	11.01	0.35	0.38	0.03	5.86	2.74	0.81	0.06	0.85	0.23	71.74	< 0.01	0.63	5.10	99.79	5	3020	65.5	13.0
TB-6-FC	299	297	6.19	0.04	1.61	0.02	2.66	1.03	0.50	0.09	1.76	0.04	81.88	< 0.01	0.18	2.24	98.24	< 1	396	33.5	4.5
FC-02-01-9.9M	299	297	14.04	0.12	0.66	< 0.01	3.54	3.52	1.20	0.08	0.46	0.15	69.96	< 0.01	0.60	4.46	98.79	< 1	1225	116.5	14.5
FC-02-01-10.6M	299	297	7.08	0.06	16.30	0.01	5.73	1.91	9.25	0.47	0.25	0.12	33.93	0.04	0.32	24.25	99.72	< 1	543	66.0	9.5
FC-02-01-23.1M	299	297	13.88	0.11	0.38	< 0.01	2.89	3.80	0.98	0.07	0.37	0.11	71.89	< 0.01	0.62	3.83	98.93	< 1	991	115.0	15.5
FC-02-01-66.4M	299	297	9.81	0.05	1.43	0.01	4.59	1.54	1.99	0.11	0.68	0.05	71.80	< 0.01	0.57	5.45	98.08	< 1	461	65.0	11.0
FC-02-01-92.2M	299	297	10.21	0.08	1.06	0.01	4.30	1.76	1.44	0.15	0.94	0.06	73.29	< 0.01	0.44	4.88	98.62	< 1	639	62.0	8.0
FC-02-03-44.6M	299	297	13.37	0.03	7.76	0.05	9.24	0.38	7.43	0.10	0.10	0.52	39.41	0.10	0.71	19.10	98.30	< 1	206	118.0	29.0
FC-02-03-60.3M	299	297	12.21	0.04	8.27	0.05	12.21	0.73	8.57	0.24	1.10	0.29	37.50	0.03	2.16	16.00	99.40	< 1	293	47.5	55.5
FC-02-03-73.5M	299	297	23.57	0.16	0.34	0.01	8.08	5.74	2.50	0.12	0.89	0.18	48.28	0.01	0.73	9.22	99.83	< 1	1600	158.0	18.5
FC-02-03-83.7M	299	297	12.23	0.03	9.19	0.05	10.51	0.27	6.45	0.30	2.01	0.25	40.31	0.03	2.07	15.85	99.55	< 1	56.0	40.0	51.0
FC-02-03-89.7M	299	297	9.13	0.07	1.13	0.01	5.31	1.50	2.10	0.21	1.28	0.05	71.23	0.01	0.48	6.05	98.56	< 1	558	55.5	11.0
FC-02-05-15.3M	299	297	6.10	0.05	0.15	0.01	2.68	1.24	0.98	0.11	0.40	0.05	83.09	< 0.01	0.41	3.92	99.19	< 1	325	60.0	2.5
FC-02-05-27.5M	299	297	7.33	0.03	0.10	< 0.01	12.29	0.69	3.71	0.20	0.89	0.04	63.99	< 0.01	0.35	9.81	99.43	< 1	171.0	55.0	16.0
FC-02-05-43.8M	299	297	9.67	0.05	0.13	< 0.01	16.32	1.02	3.48	0.18	1.03	0.05	55.88	< 0.01	0.51	10.75	99.07	< 1	246	72.0	22.5
FC-02-05-60.8M	299	297	14.40	0.10	0.12	0.01	9.87	2.92	3.28	0.16	0.63	0.09	58.65	< 0.01	0.67	7.34	98.24	< 1	865	77.5	12.5
FC-02-05-77.1M	299	297	7.26	0.07	3.90	0.01	2.67	1.77	1.79	0.13	0.29	0.01	74.05	< 0.01	0.33	6.20	98.48	< 1	581	68.0	4.5
FC-02-05-89.65M	299	297	11.00	0.07	11.16	< 0.01	9.07	2.89	5.42	0.31	0.43	0.08	38.80	0.02	0.51	19.89	99.65	< 1	562	65.0	18.5
FC-02-05-117.5M	299	297	7.39	0.05	0.11	0.02	1.90	1.88	0.72	0.07	0.38	0.05	83.62	< 0.01	0.21	2.34	98.74	< 1	385	42.5	4.0
FC-02-05-137.7M	299	297	11.45	0.10	0.30	0.01	4.28	3.05	1.80	0.11	0.44	0.13	70.98	< 0.01	0.50	5.29	98.44	< 1	931	83.5	8.5
FC-02-05-153.3M	299	297	10.88	0.04	3.09	0.03	7.63	1.68	4.15	0.15	0.69	0.12	59.15	< 0.01	1.00	9.60	98.21	< 1	383	69.5	23.0

CERTIFICATION:

* SAMPLE "02115" WAS RE-WEIGHED, FUSED & ANALYZED. LOW TOTALS OBTAINED BOTH TIMES.

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
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To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Page Number :1-B
Total Pages :2
Certificate Date: 09-AUG-2002
Invoice No. : I0221520
P.O. Number :
Account :PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm	Ni ppm	Pb ppm	Pr ppm	Rb ppm
02111	299	297	180	< 0.1	990	0.9	0.6	0.3	1	1.3	< 1	0.1	6.5	< 0.1	< 2	1	6.0	35	25	1.6	1.4
02112	299	297	130	1.9	70	3.6	2.2	1.2	14	5.5	6	0.7	34.5	0.3	< 2	10	28.0	25	25	7.9	90.0
02113	299	297	170	0.1	80	2.0	0.9	0.6	5	3.7	3	0.4	19.0	0.1	< 2	3	16.5	5	15	4.5	1.2
02114	299	297	90	1.5	175	3.3	1.7	2.8	8	3.8	1	0.6	13.0	0.2	< 2	3	12.0	30	1830	3.0	55.8
02115	299	297	80	1.0	1100	3.2	1.9	2.2	7	3.3	< 1	0.7	11.0	0.2	6	1	9.0	60	>10000	2.4	42.0
02116	299	297	50	0.8	575	4.4	2.4	3.7	7	4.9	< 1	0.9	18.0	0.3	< 2	1	16.5	40	3170	4.3	32.4
02117	299	297	280	1.0	20	3.2	2.3	1.0	9	3.7	1	0.7	17.0	0.3	12	6	14.0	65	315	3.6	29.8
02118	299	297	200	1.1	120	5.1	2.9	2.0	12	6.6	2	1.0	35.5	0.4	10	9	27.0	80	20	7.4	35.0
02119	299	297	230	1.1	250	4.6	2.8	1.8	12	6.4	3	1.0	42.5	0.4	4	12	29.5	35	15	8.0	34.8
02120	299	297	30	0.5	150	1.2	0.7	0.3	2	1.3	< 1	0.2	6.0	< 0.1	2	1	5.5	10	40	1.5	12.6
02121	299	297	230	1.4	1660	2.6	1.7	0.9	9	3.2	1	0.6	17.5	0.2	8	6	14.5	45	140	3.9	42.0
02122	299	297	120	3.0	245	2.7	1.5	1.0	20	4.3	2	0.5	38.5	0.2	< 2	10	27.5	45	155	8.2	91.4
02123	299	297	180	3.0	< 5	3.9	2.2	0.9	18	5.1	4	0.8	30.5	0.3	< 2	16	25.5	5	90	7.3	156.0
02124	299	297	140	2.2	5	3.5	2.0	0.8	18	5.0	4	0.7	29.0	0.2	< 2	18	24.0	5	20	6.9	171.0
02125	299	297	210	1.1	60	4.5	2.8	1.4	13	5.4	3	1.0	26.5	0.3	8	13	22.0	30	15	5.8	37.6
TB-1-SCR	299	297	290	0.1	5	1.8	0.9	0.6	1	3.2	2	0.3	22.5	< 0.1	< 2	2	18.0	< 5	5	5.0	2.2
TB-2-SCR	299	297	210	1.4	15	4.8	3.0	1.7	10	5.7	2	1.0	26.5	0.4	12	6	23.0	40	15	6.1	40.0
TB-3-F6	299	297	390	4.9	20	5.2	2.9	2.3	22	6.4	6	1.1	27.5	0.3	< 2	49	29.5	135	35	7.5	74.2
TB-4-FC	299	297	150	0.7	840	2.0	1.1	0.3	4	2.4	1	0.4	11.0	0.1	< 2	1	9.5	25	50	2.6	35.6
TB-5-FC	299	297	220	2.6	360	4.2	2.5	1.2	17	5.4	6	0.9	34.0	0.3	8	12	28.0	50	750	8.0	125.5
TB-6-FC	299	297	180	0.9	10	1.6	0.9	0.5	6	2.3	4	0.3	18.5	0.1	< 2	4	13.5	5	20	4.0	45.6
FC-02-01-9.9M	299	297	110	3.1	10	4.9	2.8	1.5	17	7.7	4	1.0	59.0	0.3	< 2	20	48.0	30	10	13.6	168.5
FC-02-01-10.6M	299	297	70	1.4	< 5	3.8	2.1	1.2	9	5.1	1	0.8	33.0	0.2	< 2	10	28.0	20	60	7.6	79.4
FC-02-01-23.1M	299	297	110	3.7	55	5.3	3.0	1.2	18	7.7	3	1.1	57.5	0.4	< 2	15	47.0	35	75	13.4	181.5
FC-02-01-66.4M	299	297	140	1.5	20	3.1	1.9	1.0	13	4.5	8	0.6	31.5	0.3	< 2	10	26.0	20	5	7.5	72.4
FC-02-01-92.2M	299	297	160	2.2	25	3.0	1.7	1.0	12	4.5	6	0.6	31.0	0.2	< 2	7	25.0	20	5	7.2	81.0
FC-02-03-44.6M	299	297	430	4.3	45	4.0	2.0	2.7	16	8.7	3	0.7	57.0	0.2	< 2	9	62.5	155	15	15.8	14.4
FC-02-03-60.3M	299	297	430	1.0	75	4.8	2.5	1.9	17	6.2	4	0.9	22.5	0.3	2	26	26.0	150	5	6.4	32.6
FC-02-03-73.5M	299	297	140	4.8	20	6.9	4.0	2.3	29	10.9	4	1.4	86.0	0.5	< 2	15	68.0	40	15	19.6	240
FC-02-03-83.7M	299	297	440	0.3	130	4.0	2.1	1.4	16	5.1	4	0.8	19.0	0.2	2	22	21.5	95	20	5.3	8.2
FC-02-03-89.7M	299	297	150	1.5	30	2.7	1.6	0.9	12	3.8	6	0.5	27.0	0.2	< 2	8	22.0	20	5	6.4	71.2
FC-02-05-15.3M	299	297	140	1.6	55	4.5	2.9	0.6	9	5.1	2	1.0	30.0	0.4	< 2	10	26.0	20	< 5	7.0	67.2
FC-02-05-27.5M	299	297	90	0.7	385	2.6	1.7	0.3	11	3.5	2	0.6	27.0	0.2	< 2	7	21.5	20	60	6.2	29.6
FC-02-05-43.8M	299	297	80	1.0	90	3.4	2.2	0.4	14	4.8	4	0.8	35.0	0.3	< 2	16	30.5	20	155	8.8	47.4
FC-02-05-60.8M	299	297	110	2.9	25	3.6	2.2	0.8	20	5.3	7	0.8	38.5	0.3	< 2	13	31.0	30	20	8.9	143.0
FC-02-05-77.1M	299	297	140	1.6	10	2.6	1.4	0.9	8	4.4	8	0.5	33.0	0.2	< 2	6	26.0	5	35	7.5	83.8
FC-02-05-89.65M	299	297	110	2.1	15	3.5	2.1	1.3	15	5.0	4	0.7	31.5	0.3	< 2	10	26.5	30	5	7.4	125.5
FC-02-05-117.5M	299	297	200	1.6	< 5	1.6	0.9	0.5	8	2.5	5	0.3	21.5	0.1	< 2	4	16.0	5	25	4.7	92.4
FC-02-05-137.7M	299	297	130	2.9	25	3.5	2.1	1.1	16	5.3	10	0.7	41.5	0.3	< 2	11	33.5	15	15	9.6	150.0
FC-02-05-153.3M	299	297	280	1.7	20	4.6	2.6	1.0	14	5.9	13	0.9	35.5	0.4	< 2	13	30.5	55	20	8.3	80.0

CERTIFICATION:

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ALS Chemex
Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
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To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 09-AUG-2002
Invoice No. : I0221520
P.O. Number :
Account :PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
02111	299	297	1.3	6	6.6	< 0.5	0.1	< 1	< 0.5	< 0.1	< 0.5	5	1	5.5	0.6	25	21.5
02112	299	297	5.5	15	66.1	0.5	0.7	14	< 0.5	0.3	3.0	55	3	18.5	1.9	80	204
02113	299	297	3.4	2	5.3	< 0.5	0.5	3	< 0.5	0.1	0.5	5	1	8.5	0.6	30	104.5
02114	299	297	2.9	4	306	< 0.5	0.6	3	< 0.5	0.2	0.5	40	3	20.5	1.5	2820	36.0
02115	299	297	2.6	9	74.0	< 0.5	0.5	1	< 0.5	0.3	0.5	50	3	21.5	1.6	>10000	31.0
02116	299	297	4.3	4	274	< 0.5	0.8	3	< 0.5	0.3	0.5	40	4	28.5	1.9	7960	26.5
02117	299	297	3.0	8	71.3	< 0.5	0.5	3	< 0.5	0.3	5.0	780	8	28.5	2.0	880	46.5
02118	299	297	5.7	37	107.0	0.5	0.9	5	< 0.5	0.4	7.0	335	5	35.5	2.2	290	76.0
02119	299	297	5.8	15	53.8	1.0	0.9	7	< 0.5	0.4	5.5	280	4	32.0	2.3	240	104.5
02120	299	297	1.2	< 1	9.6	< 0.5	0.2	1	< 0.5	< 0.1	1.5	20	2	7.5	0.6	30	14.5
02121	299	297	2.9	12	22.9	0.5	0.4	4	< 0.5	0.2	3.5	370	4	19.0	1.4	170	58.5
02122	299	297	4.8	8	24.7	0.5	0.5	10	< 0.5	0.1	2.5	110	7	14.5	1.3	190	77.0
02123	299	297	5.3	10	122.5	2.5	0.8	12	< 0.5	0.3	2.5	15	4	22.0	1.9	25	111.5
02124	299	297	5.1	5	77.7	2.5	0.7	11	< 0.5	0.3	5.5	15	4	19.5	1.7	35	101.5
02125	299	297	4.8	21	80.9	1.0	0.8	5	< 0.5	0.4	7.5	445	6	34.0	2.2	175	91.5
TB-1-SCR	299	297	3.3	< 1	4.7	< 0.5	0.4	1	< 0.5	0.1	< 0.5	< 5	2	9.0	0.6	5	77.0
TB-2-SCR	299	297	5.0	11	62.7	0.5	0.8	5	< 0.5	0.4	7.0	905	6	37.0	2.4	115	87.0
TB-3-F6	299	297	6.2	37	111.0	3.5	1.0	5	< 0.5	0.4	1.5	280	4	28.0	2.3	115	215
TB-4-FC	299	297	2.0	12	16.2	< 0.5	0.4	3	< 0.5	0.1	1.5	5	2	12.0	0.8	90	41.0
TB-5-FC	299	297	5.4	83	45.8	1.0	0.8	11	7.5	0.3	5.0	210	10	24.0	2.1	3000	202
TB-6-FC	299	297	2.4	< 1	66.6	< 0.5	0.3	6	< 0.5	0.1	1.0	20	< 1	9.0	0.9	30	130.0
FC-02-01-9.9M	299	297	8.2	2	49.6	1.5	1.0	14	< 0.5	0.4	2.0	80	3	25.5	2.2	40	117.5
FC-02-01-10.6M	299	297	5.4	1	310	0.5	0.7	7	< 0.5	0.3	1.5	40	2	21.0	1.6	90	74.5
FC-02-01-23.1M	299	297	8.5	38	40.0	1.0	1.0	14	< 0.5	0.4	2.5	85	7	26.5	2.3	75	124.5
FC-02-01-66.4M	299	297	4.7	2	79.0	0.5	0.6	12	< 0.5	0.3	2.5	60	2	17.5	1.7	70	292
FC-02-01-92.2M	299	297	4.6	1	72.1	0.5	0.6	11	< 0.5	0.2	1.5	45	1	17.0	1.5	50	187.0
FC-02-03-44.6M	299	297	10.7	1	960	0.5	0.9	7	< 0.5	0.2	5.5	130	1	20.0	1.6	90	128.0
FC-02-03-60.3M	299	297	5.9	1	262	2.0	0.9	2	< 0.5	0.3	1.5	300	5	24.0	1.9	120	164.5
FC-02-03-73.5M	299	297	11.9	3	132.0	1.5	1.4	21	0.5	0.5	4.5	105	4	37.0	3.3	85	141.0
FC-02-03-83.7M	299	297	4.9	1	297	1.5	0.8	2	< 0.5	0.3	0.5	305	2	21.0	1.7	115	134.5
FC-02-03-89.7M	299	297	3.8	1	104.0	0.5	0.5	11	< 0.5	0.2	2.0	55	1	15.5	1.5	70	203
FC-02-05-15.3M	299	297	5.0	14	24.9	0.5	0.8	6	< 0.5	0.4	4.5	135	7	32.0	2.4	30	80.5
FC-02-05-27.5M	299	297	3.6	24	62.3	0.5	0.5	8	< 0.5	0.2	1.5	45	14	15.5	1.5	290	77.5
FC-02-05-43.8M	299	297	5.3	23	71.1	1.5	0.7	12	< 0.5	0.3	2.0	60	12	18.5	2.0	295	151.0
FC-02-05-60.8M	299	297	5.5	56	55.3	1.0	0.7	13	< 0.5	0.3	3.5	80	8	20.5	2.0	140	261
FC-02-05-77.1M	299	297	4.8	5	93.3	0.5	0.6	14	< 0.5	0.1	2.0	25	3	15.0	1.3	40	299
FC-02-05-89.65M	299	297	4.9	2	216	0.5	0.7	10	< 0.5	0.3	1.5	65	2	20.0	1.7	55	150.0
FC-02-05-117.5M	299	297	2.6	1	26.3	< 0.5	0.3	8	< 0.5	0.1	1.5	15	2	9.0	0.8	55	164.5
FC-02-05-137.7M	299	297	5.8	4	41.1	0.5	0.7	17	1.5	0.3	3.5	50	4	19.5	1.8	75	362
FC-02-05-153.3M	299	297	5.9	24	134.5	1.0	0.8	16	< 0.5	0.4	2.5	95	4	25.5	2.3	100	464

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ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
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PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 09-AUG-2002
Invoice No. : I0221520
P.O. Number :
Account :PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Al2O3 % XRF	BaO % XRF	CaO % XRF	Cr2O3 % XRF	Fe2O3 % XRF	K2O % XRF	MgO % XRF	MnO % XRF	Na2O % XRF	P2O5 % XRF	SiO2 % XRF	SrO % XRF	TiO2 % XRF	LOI % XRF	TOTAL %	Ag ppm	Ba ppm	Ce ppm	Co ppm
FC-02-06-26.6M	299	297	11.09	0.06	2.84	0.03	7.89	1.76	4.02	0.15	0.70	0.11	59.97	0.01	0.97	9.87	99.47	< 1	383	69.0	29.0
FC-02-06-42.7M	299	297	8.95	0.06	0.12	0.01	6.27	1.87	2.04	0.12	0.33	0.04	74.18	0.01	0.50	4.52	99.02	8	489	67.5	9.0
FC-02-06-61.1M	299	297	15.03	0.05	2.62	0.06	15.74	1.34	10.80	0.32	0.49	0.33	38.08	0.01	2.66	11.88	99.41	14	374	60.0	81.5
FC-02-06-77.8M	299	297	9.59	0.06	6.50	0.01	4.50	2.23	2.93	0.16	0.48	0.04	60.36	0.01	0.38	11.31	98.56	1	561	65.5	8.5
FC-02-06-101.4M	299	297	13.25	0.06	1.09	0.01	4.32	2.73	1.89	0.09	1.16	0.11	68.63	< 0.01	0.59	5.94	99.87	< 1	457	75.5	11.5
FC-02-06-115.0M	299	297	8.79	0.04	0.68	0.02	2.17	1.69	1.16	0.06	1.14	0.06	80.16	< 0.01	0.25	3.15	99.37	< 1	283	45.0	4.0
FC-02-06-132.9M	299	297	11.06	0.06	0.21	0.01	7.17	2.27	3.05	0.12	0.28	0.09	69.60	< 0.01	0.47	5.41	99.80	< 1	460	68.0	18.0
ACE-02-03-119.0M	299	297	19.70	0.09	0.65	0.01	8.07	3.57	2.47	0.08	1.79	0.15	57.45	0.01	0.84	4.18	99.06	< 1	832	99.5	26.5
ACE-02-03-122.0M	299	297	19.86	0.10	0.56	0.01	9.48	3.41	2.54	0.08	1.72	0.19	56.66	0.01	1.29	4.04	99.95	< 1	821	117.5	34.0
ACE-02-04-17.1M	299	297	14.52	0.13	1.69	0.01	4.62	3.31	1.39	0.10	2.58	0.26	67.60	0.01	1.05	2.52	99.79	< 1	1225	104.0	17.5
ACE-02-04-28.0M	299	297	15.37	0.13	0.83	0.01	6.59	3.02	1.91	0.08	1.95	0.27	65.31	0.02	1.22	2.59	99.30	< 1	1190	97.0	22.5
ACE-02-04-37.3M	299	297	15.50	0.05	6.26	0.01	5.51	0.81	2.77	0.05	7.63	0.33	53.43	0.06	0.67	5.32	98.40	< 1	398	70.5	19.5
ACE-02-04-50.0M	299	297	16.90	0.26	2.94	0.01	6.54	1.12	1.80	0.07	7.82	0.05	54.51	0.05	0.60	6.12	98.79	< 1	2360	78.0	18.5
ACE-02-04-71.3M	299	297	17.82	0.20	6.97	0.01	5.55	3.70	3.05	0.07	4.32	0.11	49.48	0.05	0.63	6.13	98.09	< 1	1905	72.0	25.0
ACE-02-04-83.0M	299	297	10.20	0.14	4.01	0.01	4.04	0.66	1.56	0.13	5.28	0.60	68.81	0.03	0.35	3.12	98.94	< 1	1190	40.0	12.0
ACE-02-04-98.1M	299	297	19.09	0.16	0.59	0.02	8.31	3.46	2.37	0.10	1.52	0.14	58.86	0.01	1.01	3.85	99.49	< 1	1490	111.0	30.5
ACE-02-04-112.0M	299	297	20.08	0.16	0.48	0.01	9.34	3.60	2.44	0.08	1.47	0.11	56.82	0.01	1.10	3.73	99.43	< 1	1440	110.0	35.0

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.o: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 09-AUG-200
Invoice No. : I0221520
P.O. Number :
Account : PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Cr ppm	Cs ppm	Cu ppm	Dy ppm	Er ppm	Eu ppm	Ga ppm	Gd ppm	Hf ppm	Ho ppm	La ppm	Lu ppm	Mo ppm	Nb ppm	Nd ppm	Ni ppm	Pb ppm	Pr ppm	Rb ppm
FC-02-06-26.6M	299	297	280	1.5	20	4.3	2.4	1.0	14	5.7	12	0.8	35.5	0.3	< 2	13	30.0	60	5	8.3	79.6
FC-02-06-42.7M	299	297	160	1.7	35	2.8	1.7	0.6	11	4.2	13	0.6	33.5	0.3	< 2	8	27.5	10	35	7.8	88.2
FC-02-06-61.1M	299	297	570	1.1	150	5.1	2.6	1.6	21	6.8	5	1.0	28.5	0.3	< 2	34	30.5	210	55	7.7	56.0
FC-02-06-77.8M	299	297	150	1.7	25	3.3	1.8	1.0	12	5.0	7	0.7	31.5	0.2	2	7	26.0	15	10	7.3	100.0
FC-02-06-101.4M	299	297	150	2.2	5	3.3	2.1	1.0	18	5.0	8	0.7	38.0	0.3	< 2	12	31.0	25	5	8.9	124.0
FC-02-06-115.0M	299	297	210	1.4	10	1.6	0.9	0.6	9	2.8	5	0.3	23.0	0.1	< 2	5	17.0	5	15	5.0	74.2
FC-02-06-132.9M	299	297	150	2.0	50	3.0	1.8	0.8	15	4.6	6	0.6	34.0	0.3	< 2	9	27.0	35	< 5	7.8	104.5
ACE-02-03-119.0M	299	297	180	3.7	35	5.6	3.2	1.6	27	7.8	5	1.1	52.5	0.4	< 2	17	43.0	45	10	12.1	160.0
ACE-02-03-122.0M	299	297	160	3.5	65	8.2	5.0	2.1	27	10.5	7	1.8	61.0	0.7	< 2	22	53.0	45	5	14.7	152.0
ACE-02-04-17.1M	299	297	170	4.7	20	6.5	4.2	1.9	20	8.6	16	1.4	51.0	0.7	2	17	45.0	25	< 5	12.3	131.0
ACE-02-04-28.0M	299	297	130	2.3	20	7.0	4.3	1.9	21	8.8	9	1.5	47.0	0.6	< 2	19	42.5	25	< 5	11.8	121.5
ACE-02-04-37.3M	299	297	160	3.3	15	3.1	1.6	1.0	16	5.3	4	0.6	37.0	0.2	4	12	30.5	35	35	8.8	42.6
ACE-02-04-50.0M	299	297	160	2.2	20	3.2	2.0	1.1	23	5.1	2	0.7	45.5	0.2	56	11	33.5	50	5	9.7	55.8
ACE-02-04-71.3M	299	297	150	10.0	55	2.4	1.3	1.0	23	4.7	3	0.5	43.0	0.1	2	12	31.0	40	10	9.1	153.0
ACE-02-04-83.0M	299	297	190	2.8	30	4.1	2.2	0.9	13	4.5	1	0.8	22.5	0.2	134	7	19.5	60	55	5.2	39.2
ACE-02-04-98.1M	299	297	210	3.8	30	6.2	3.7	1.9	27	8.5	6	1.3	55.5	0.5	< 2	21	46.5	50	5	13.2	156.5
ACE-02-04-112.0M	299	297	160	4.0	60	6.0	3.4	1.8	28	8.1	5	1.2	56.0	0.5	< 2	21	46.0	50	20	13.0	165.5

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PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 09-AUG-200
Invoice No. : I0221520
P.O. Number :
Account : PFD

* PLEASE NOTE

CERTIFICATE OF ANALYSIS A0221520

SAMPLE	PREP CODE		Sm ppm	Sn ppm	Sr ppm	Ta ppm	Tb ppm	Th ppm	Tl ppm	Tm ppm	U ppm	V ppm	W ppm	Y ppm	Yb ppm	Zn ppm	Zr ppm
FC-02-06-26.6M	299	297	5.6	23	123.5	1.0	0.8	16	< 0.5	0.3	2.5	100	5	23.0	2.1	105	412
FC-02-06-42.7M	299	297	4.5	31	46.2	0.5	0.5	16	< 0.5	0.2	2.5	35	4	15.5	1.6	230	428
FC-02-06-61.1M	299	297	6.5	13	97.6	2.5	1.0	4	< 0.5	0.3	1.5	365	25	25.5	1.9	535	185.0
FC-02-06-77.8M	299	297	4.8	7	152.5	0.5	0.7	12	< 0.5	0.2	2.0	40	3	19.0	1.5	75	230
FC-02-06-101.4M	299	297	5.4	3	87.1	0.5	0.7	15	< 0.5	0.3	3.5	70	2	19.5	1.9	95	251
FC-02-06-115.0M	299	297	3.0	1	58.7	< 0.5	0.3	10	< 0.5	0.1	2.0	25	1	9.0	0.9	45	159.0
FC-02-06-132.9M	299	297	4.9	14	23.6	0.5	0.6	13	1.5	0.2	4.0	50	4	17.0	1.6	140	195.0
ACE-02-03-119.0M	299	297	8.0	3	177.0	1.5	1.0	15	< 0.5	0.4	2.5	115	< 1	31.5	2.9	130	137.0
ACE-02-03-122.0M	299	297	10.3	3	153.0	1.5	1.5	16	< 0.5	0.7	3.5	160	< 1	49.5	4.4	150	225
ACE-02-04-17.1M	299	297	8.6	2	150.0	1.5	1.2	18	< 0.5	0.6	4.0	100	1	41.0	3.8	65	551
ACE-02-04-28.0M	299	297	8.5	3	139.5	1.5	1.3	13	< 0.5	0.6	3.0	115	1	41.0	3.7	120	306
ACE-02-04-37.3M	299	297	5.7	< 1	447	0.5	0.7	12	< 0.5	0.1	6.0	200	< 1	16.5	1.3	390	114.5
ACE-02-04-50.0M	299	297	5.7	2	463	0.5	0.6	12	< 0.5	0.3	8.5	510	3	20.0	1.6	75	63.0
ACE-02-04-71.3M	299	297	5.1	3	395	0.5	0.5	12	< 0.5	0.1	3.0	150	1	12.5	1.1	65	82.5
ACE-02-04-83.0M	299	297	4.2	1	265	< 0.5	0.7	7	< 0.5	0.3	6.5	160	1	26.0	1.6	165	54.0
ACE-02-04-98.1M	299	297	8.8	3	141.0	1.5	1.2	15	< 0.5	0.5	2.5	125	1	36.5	3.3	160	184.0
ACE-02-04-112.0M	299	297	8.6	3	135.5	1.5	1.1	15	< 0.5	0.5	3.0	135	1	34.0	3.1	165	158.0

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Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Page Number :1-A
Total Pages :2
Certificate Date: 13-AUG-200
Invoice No. : I0221518
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0221518

SAMPLE	PREP CODE		Weight Kg	Au ppb ICP	Pt ppb ICP	Pd ppb ICP	Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)
02111	413	402	0.80	20	< 5	< 2	2.46	0.99	182.0	30.0	0.10	21.5	0.58	0.20	20.8	873.0	80	0.05	1985.0	15.45	5.50
02112	413	402	1.06	4	< 5	< 2	0.20	5.86	14.4	455.5	1.05	0.67	0.32	0.02	73.0	25.6	65	1.95	81.9	4.77	15.65
02113	413	402	0.58	< 2	< 5	< 2	0.08	1.48	20.0	8.5	0.15	1.01	0.14	0.06	43.2	25.7	128	0.10	85.4	4.79	5.15
02114	413	402	0.30	14	< 5	< 2	2.38	2.30	7.4	307.0	0.55	4.82	13.50	8.88	25.2	21.1	54	1.60	248.0	8.27	6.85
02115	413	402	0.16	168	10	< 2	56.5	1.58	7.0	188.0	0.65	99.6	2.60	>500	18.10	78.5	44	1.15	1960.0	11.30	5.95
02116	413	402	0.44	6	< 5	< 2	3.32	1.58	2.4	181.5	0.40	2.79	14.50	29.9	32.8	16.9	34	0.85	819.0	9.76	5.10
02117	413	402	0.72	4	< 5	12	0.40	2.84	10.8	329.0	0.50	6.96	2.00	4.58	22.3	12.6	180	1.00	27.0	5.15	8.90
02118	413	402	0.36	4	5	8	0.12	4.34	2.0	499.0	1.25	4.21	1.15	1.02	52.0	35.2	130	1.05	132.0	6.35	12.85
02119	413	402	0.52	4	5	8	0.22	3.68	1.4	413.0	0.70	6.58	0.63	0.22	57.3	19.0	144	0.95	252.0	6.71	10.40
02120	413	402	0.68	12	20	4	1.14	0.75	4.6	16.5	0.30	5.90	0.30	0.18	10.40	87.7	59	0.50	2150	>25.0	2.90
02121	413	402	0.88	< 2	< 5	6	0.12	3.10	2.6	749.2	1.00	0.64	0.34	0.56	30.9	9.2	126	1.50	102.3	3.64	9.85
02122	413	402	0.30	4	< 5	< 2	0.20	6.86	2.4	55.0	2.00	2.16	0.03	0.14	60.2	27.7	77	3.20	510.0	12.90	19.85
02123	413	402	0.52	4	< 5	< 2	0.04	7.78	5.0	854.0	1.55	0.14	0.19	0.04	66.4	3.8	59	3.20	5.0	1.74	18.70
02124	413	402	0.28	2	< 5	< 2	0.06	7.26	6.6	3000	1.30	0.26	0.23	0.18	66.1	4.5	50	2.35	6.0	1.42	19.05
02125	413	402	1.00	2	5	6	0.14	4.65	3.8	507.1	1.15	5.21	1.10	0.08	44.7	13.4	134	1.10	67.2	6.67	13.90
TB-1-SCR	413	402	1.20	< 2	5	< 2	0.02	0.18	9.0	34.5	< 0.05	0.48	0.04	0.02	49.0	0.9	176	0.05	8.0	0.53	1.45
TB-2-SCR	413	402	1.00	< 2	5	8	0.02	3.17	6.2	812.0	0.95	0.54	1.45	0.04	44.6	8.9	149	1.40	15.6	3.48	10.10
TB-3-F6	413	402	1.44	< 2	< 5	< 2	0.06	7.99	14.4	2820	1.65	1.65	1.75	< 0.02	59.0	36.6	252	4.95	20.8	9.16	22.05
TB-4-FC	413	402	0.94	12	5	< 2	1.56	1.93	52.8	194.5	0.40	3.04	0.05	0.24	24.9	47.8	105	0.70	1545.0	10.30	4.50
TB-5-FC	413	402	1.14	14	< 5	< 2	6.98	5.59	260	78.0	1.35	6.45	0.24	10.45	58.7	11.8	124	2.55	386.0	3.79	16.15
TB-6-FC	413	402	0.88	< 2	< 5	< 2	0.06	3.23	3.0	378.0	0.45	0.20	1.05	0.06	30.5	4.0	72	0.80	9.4	1.68	6.60
FC-02-01-9.9M	413	402	0.80	2	< 5	< 2	0.06	7.10	25.6	1090.0	2.00	0.14	0.46	0.02	117.5	14.2	46	3.10	11.0	2.22	18.70
FC-02-01-10.6M	413	402	0.68	4	< 5	< 2	0.14	3.61	19.8	521.7	1.20	0.25	10.50	0.16	64.8	8.9	35	1.30	4.0	3.32	9.15
FC-02-01-23.1M	413	402	0.60	2	< 5	< 2	0.30	7.33	49.0	706.4	1.65	0.32	0.25	0.10	117.0	15.1	45	3.60	61.7	1.88	20.35
FC-02-01-66.4M	413	402	0.92	< 2	< 5	< 2	0.08	5.58	16.2	471.5	1.30	0.12	1.00	0.06	67.4	11.2	71	1.50	19.0	3.12	14.85
FC-02-01-92.2M	413	402	0.82	< 2	< 5	< 2	0.08	5.45	8.4	640.0	1.20	0.21	0.70	0.02	63.8	8.3	84	2.20	24.4	2.78	13.30
FC-02-03-44.6M	413	402	0.50	< 2	< 5	< 2	0.12	7.33	2.6	190.0	1.50	0.14	5.40	0.12	116.0	26.4	201	4.20	44.0	5.51	16.40
FC-02-03-60.3M	413	402	0.92	< 2	< 5	< 2	0.14	6.26	81.8	296.5	0.95	0.04	5.50	0.14	49.2	49.8	218	0.95	68.9	7.60	18.30
FC-02-03-73.5M	413	402	0.66	< 2	< 5	< 2	0.06	12.15	14.2	1474.0	3.30	0.10	0.23	0.02	162.5	19.1	65	4.85	23.4	5.15	31.65
FC-02-03-83.7M	413	402	0.98	< 2	5	6	0.16	6.16	46.2	53.5	0.55	0.03	5.90	0.14	41.3	43.2	187	0.25	116.4	6.39	16.40
FC-02-03-89.7M	413	402	0.72	< 2	< 5	< 2	0.06	5.18	9.2	549.1	1.30	0.07	0.78	0.12	57.5	9.4	62	1.50	31.8	3.61	13.35
FC-02-05-15.3M	413	402	0.58	< 2	< 5	2	0.14	3.06	30.6	328.0	0.40	0.17	0.09	< 0.02	61.3	2.3	56	1.55	53.5	1.74	9.35
FC-02-05-27.5M	413	402	0.52	32	< 5	< 2	1.48	3.66	50.4	163.5	0.30	2.45	0.06	0.88	54.4	14.0	49	0.65	415.0	7.85	11.30
FC-02-05-43.8M	413	402	0.84	< 2	< 5	< 2	1.28	4.80	31.4	225.0	0.40	2.68	0.07	0.40	67.3	16.2	47	0.90	81.2	10.00	14.20
FC-02-05-60.8M	413	402	0.86	< 2	< 5	< 2	0.18	7.73	46.4	843.6	1.20	0.66	0.05	< 0.02	79.7	12.3	61	2.95	24.6	6.31	20.65
FC-02-05-77.1M	413	402	0.86	< 2	< 5	< 2	0.16	3.91	8.8	543.8	0.75	0.45	2.60	0.08	67.8	4.1	63	1.55	10.2	1.74	8.60
FC-02-05-89.65M	413	402	0.50	< 2	< 5	< 2	0.02	5.80	22.6	570.8	1.50	0.13	7.50	0.10	71.0	17.8	53	2.10	17.6	5.74	16.10
FC-02-05-117.5M	413	402	0.78	< 2	< 5	< 2	0.18	3.91	5.4	396.0	1.05	0.43	0.07	0.14	46.5	4.0	67	1.65	3.2	1.23	8.50
FC-02-05-137.7M	413	402	0.54	< 2	< 5	< 2	0.08	6.19	15.6	909.8	1.55	0.14	0.20	0.20	85.8	8.0	47	2.95	24.0	2.82	16.75
FC-02-05-153.3M	413	402	0.56	2	< 5	< 2	0.10	6.67	33.8	408.0	0.90	0.17	2.00	0.06	77.2	21.4	121	1.60	16.8	4.78	16.65

CERTIFICATION: [signature]

ALS Chemex

Aurora Laboratory Services Ltd.
Analytical Chemists * Geochemists * Registered Assayers
212 Brooksbank Ave., North Vancouver
British Columbia, Canada V7J 2C1
PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 13-AUG-200
Invoice No. : I0221518
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0221518

SAMPLE	PREP CODE	Ge ppm (ICP)	Hf ppm	In ppm	K % (ICP)	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm
02111	9413 9402	0.30	0.1	3.47	0.03	9.5	10.8	0.13	290	1.10	< 0.01	1.3	66.1	190	30.5	1.4	0.004	>10.00	0.80	12
02112	9413 9402	0.20	0.4	0.145	1.54	37.0	24.4	1.01	700	0.75	1.04	5.4	33.0	430	22.0	85.8	0.004	1.01	0.15	1
02113	9413 9402	0.15	0.8	0.060	0.01	18.5	8.8	0.78	350	0.60	< 0.01	2.0	9.0	320	3.0	1.0	0.004	1.58	0.25	1
02114	9413 9402	0.20	0.6	0.110	0.84	14.0	11.6	5.12	>10000	0.55	0.05	0.6	41.6	180	2130	56.9	0.004	1.04	1.90	3
02115	9413 9402	0.45	0.5	5.35	0.61	10.0	7.4	1.13	>10000	0.90	0.04	0.6	85.3	290	>10000	44.0	0.004	4.09	28.55	91
02116	9413 9402	0.25	0.6	0.175	0.52	18.0	7.4	4.42	>10000	0.50	0.04	0.6	62.0	180	3630	31.7	0.002	0.65	4.30	9
02117	9413 9402	0.15	0.1	0.855	0.40	16.0	24.4	1.44	3210	11.80	0.01	1.6	88.7	1990	266	27.0	0.010	0.04	0.30	2
02118	9413 9402	0.20	0.1	0.255	0.55	35.5	23.0	0.83	>10000	9.35	0.76	3.0	103.0	5380	24.0	30.5	0.010	0.03	0.30	1
02119	9413 9402	0.15	< 0.1	0.305	0.48	40.5	23.4	0.91	3320	3.60	0.26	2.4	43.0	2930	23.0	27.2	0.008	0.08	0.20	1
02120	9413 9402	0.50	0.5	0.040	0.21	5.5	2.8	0.49	1165	5.85	0.02	1.1	301	160	106.5	12.8	0.008	>10.00	0.40	4
02121	9413 9402	0.10	0.1	0.160	0.74	20.0	22.2	0.75	715	8.15	0.04	2.8	52.1	1390	21.0	37.2	0.008	0.15	0.20	1
02122	9413 9402	0.25	1.7	0.100	1.67	31.0	47.0	0.38	875	1.00	0.16	3.0	54.3	300	12.0	90.6	0.002	4.43	0.30	4
02123	9413 9402	0.15	2.3	0.075	2.78	35.5	26.8	0.36	240	0.65	2.56	14.1	8.8	750	9.0	142.0	0.002	0.01	0.20	< 1
02124	9413 9402	0.15	2.3	0.035	3.21	35.0	25.0	0.37	410	0.65	2.13	17.7	11.2	880	22.0	150.0	0.006	0.01	0.25	1
02125	9413 9402	0.20	< 0.1	0.125	0.62	27.5	26.4	1.07	3020	7.55	0.55	6.2	41.0	5220	12.0	34.7	0.014	0.03	0.20	1
TB-1-SCR	9413 9402	0.05	0.1	0.025	0.03	21.5	1.2	0.02	40	1.15	< 0.01	1.5	3.0	350	5.5	1.4	0.002	0.05	0.15	< 1
TB-2-SCR	9413 9402	0.15	< 0.1	0.120	0.75	29.0	20.6	0.82	845	12.30	0.04	2.3	52.8	6770	6.0	35.4	0.010	< 0.01	0.15	1
TB-3-F6	9413 9402	0.25	1.6	0.185	1.42	28.0	43.4	3.92	1295	0.75	1.20	11.1	164.0	1530	1.0	70.2	0.006	< 0.01	0.05	2
TB-4-FC	9413 9402	0.20	0.5	0.580	0.66	11.5	3.4	0.91	1215	1.35	0.14	0.6	30.4	90	31.0	33.5	0.002	3.78	0.45	6
TB-5-FC	9413 9402	0.20	2.8	0.625	2.24	26.5	7.0	0.45	255	6.95	0.16	5.4	56.5	1050	868	111.5	0.022	2.51	23.75	6
TB-6-FC	9413 9402	0.05	0.7	0.010	0.85	16.5	2.6	0.28	500	0.65	1.20	1.3	9.8	100	23.0	36.5	0.002	0.02	0.25	< 1
FC-02-01-9.9M	9413 9402	0.20	2.4	0.045	2.97	61.5	8.2	0.67	410	0.35	0.23	10.7	37.2	660	6.5	147.0	0.004	0.94	0.40	1
FC-02-01-10.6M	9413 9402	0.20	1.0	0.025	1.52	31.5	4.4	5.05	2540	0.45	0.13	3.7	23.0	470	115.5	71.7	0.004	0.25	0.45	1
FC-02-01-23.1M	9413 9402	0.20	2.3	0.015	3.11	61.0	14.6	0.53	280	0.30	0.23	8.4	44.8	470	111.5	161.5	0.004	1.32	0.90	1
FC-02-01-66.4M	9413 9402	0.20	2.5	0.035	1.36	32.0	21.6	1.17	720	0.45	0.46	3.6	25.4	240	12.5	66.8	0.002	0.13	0.15	< 1
FC-02-01-92.2M	9413 9402	0.15	1.9	0.030	1.46	31.0	14.6	0.79	930	0.60	0.66	3.0	22.2	210	15.0	73.8	0.004	0.46	0.15	1
FC-02-03-44.6M	9413 9402	0.25	3.1	0.040	0.29	57.0	59.2	3.89	995	1.15	0.05	9.0	171.0	2150	13.0	12.3	0.004	0.16	1.95	1
FC-02-03-60.3M	9413 9402	0.20	0.8	0.075	0.58	23.0	85.2	4.60	1475	1.40	0.66	13.5	171.0	1260	4.0	29.5	0.006	0.10	0.15	1
FC-02-03-73.5M	9413 9402	0.30	3.0	0.085	4.37	91.5	19.4	1.28	780	0.60	0.62	11.1	50.2	710	6.0	219	0.006	0.01	0.20	1
FC-02-03-83.7M	9413 9402	0.20	0.5	0.065	0.20	19.0	73.4	3.34	1865	1.00	1.53	6.9	107.0	1000	8.0	6.9	0.004	0.13	0.05	1
FC-02-03-89.7M	9413 9402	0.15	2.0	0.030	1.31	27.5	13.8	1.23	1610	0.45	0.93	3.3	25.8	210	9.5	64.1	0.002	0.19	0.05	< 1
FC-02-05-15.3M	9413 9402	0.15	0.3	0.130	1.03	29.5	6.8	0.54	640	1.00	0.19	4.0	27.0	260	11.0	55.8	0.010	0.04	0.30	< 1
FC-02-05-27.5M	9413 9402	0.20	1.1	0.790	0.50	25.5	26.6	1.99	1440	0.45	0.59	3.1	19.4	140	68.5	26.1	0.002	0.74	0.35	1
FC-02-05-43.8M	9413 9402	0.20	2.1	0.380	0.77	32.0	33.8	1.74	1285	0.30	0.63	5.2	21.0	200	196.5	40.2	0.004	0.45	0.30	1
FC-02-05-60.8M	9413 9402	0.20	2.1	0.215	2.35	41.0	41.8	1.80	1115	0.50	0.41	4.8	35.8	200	19.5	129.5	0.002	0.95	0.35	< 1
FC-02-05-77.1M	9413 9402	0.15	1.9	0.035	1.49	32.5	6.2	1.01	830	0.50	0.22	3.1	8.6	110	38.5	70.6	0.004	0.20	0.20	< 1
FC-02-05-89.65M	9413 9402	0.20	1.4	0.040	2.37	34.0	7.4	3.03	2010	0.20	0.30	1.6	30.0	330	6.0	120.0	0.002	< 0.01	0.15	< 1
FC-02-05-117.5M	9413 9402	0.10	0.8	0.020	1.55	22.5	6.4	0.42	310	0.55	0.18	1.8	7.6	170	30.5	80.6	0.004	< 0.01	0.20	< 1
FC-02-05-137.7M	9413 9402	0.20	2.8	0.040	2.55	44.5	9.8	1.00	630	0.90	0.23	4.1	18.0	520	10.0	134.0	0.004	0.39	0.35	1
FC-02-05-153.3M	9413 9402	0.20	13.8	0.105	1.38	40.0	86.6	2.19	950	2.20	0.47	5.7	63.7	460	12.0	73.7	0.004	0.29	0.40	1

CERTIFICATION:

ALS Chemex

Aurora Laboratory Services Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 Brooksbank Ave., North Vancouver

British Columbia, Canada V7J 2C1

PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES

Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 13-AUG-2002
Invoice No. : I0221518
P.O. Number :
Account : PFD

CERTIFICATE OF ANALYSIS A0221518

SAMPLE	PREP CODE	Sn ppm	Sr ppm (ICP)	Ta ppm (ICP)	Te ppm (ICP)	Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
02111	9413 9402	5.6	10.0	0.20	0.75	3.4	0.07	0.06	0.8	8	2.2	5.2	30	4.2	10
02112	9413 9402	13.8	54.0	0.35	0.05	15.8	0.20	0.46	2.2	45	2.0	11.5	84	57.4	< 10
02113	9413 9402	2.0	4.2	0.10	0.05	3.8	0.06	< 0.02	0.4	11	0.6	6.5	36	8.7	< 10
02114	9413 9402	2.8	336	< 0.05	0.10	4.0	0.02	0.50	0.8	35	0.6	9.7	3170	23.1	150
02115	9413 9402	8.2	86.5	< 0.05	3.85	2.0	0.01	0.52	0.8	38	0.8	17.0	>10000	22.6	7940
02116	9413 9402	3.4	291	< 0.05	0.15	3.2	0.01	0.30	0.6	32	0.6	13.2	8820	16.4	440
02117	9413 9402	4.8	75.9	< 0.05	0.10	3.8	0.04	0.36	4.9	733	2.0	11.5	1080	7.6	50
02118	9413 9402	30.0	109.0	< 0.05	< 0.05	5.8	0.15	0.34	6.8	294	2.1	17.8	342	2.6	< 10
02119	9413 9402	9.8	46.8	< 0.05	0.05	7.2	0.06	0.34	5.1	233	1.3	11.6	248	6.7	< 10
02120	9413 9402	1.2	10.8	< 0.05	0.30	1.4	0.01	0.32	1.6	45	2.2	6.0	40	16.3	70
02121	9413 9402	9.2	23.4	< 0.05	< 0.05	5.2	0.09	0.40	3.7	359	2.0	8.8	204	11.5	< 10
02122	9413 9402	6.4	22.0	0.10	0.20	9.0	0.06	0.68	2.4	95	2.6	4.5	216	54.3	10
02123	9413 9402	9.6	122.0	1.50	< 0.05	15.0	0.15	0.54	2.4	15	3.3	10.1	24	64.4	< 10
02124	9413 9402	4.4	77.6	1.65	< 0.05	14.8	0.14	0.64	5.5	13	2.8	9.8	40	65.5	< 10
02125	9413 9402	16.8	83.0	0.10	0.05	6.8	0.27	0.38	8.0	407	2.7	14.7	186	6.7	< 10
TB-1-SCR	9413 9402	0.6	4.6	< 0.05	< 0.05	2.2	0.03	0.02	0.4	5	1.0	4.1	10	3.5	< 10
TB-2-SCR	9413 9402	8.0	63.8	< 0.05	< 0.05	6.6	0.08	0.44	7.9	860	2.5	23.3	132	3.6	< 10
TB-3-F6	9413 9402	28.6	109.0	0.05	< 0.05	5.6	1.31	0.68	1.2	229	0.3	21.6	96	62.0	< 10
TB-4-FC	9413 9402	11.4	17.0	< 0.05	0.05	3.2	0.01	0.22	1.6	10	0.6	5.0	100	11.1	< 10
TB-5-FC	9413 9402	63.2	39.4	0.20	< 0.05	6.6	0.18	13.35	4.6	178	4.4	12.5	3190	84.3	8290
TB-6-FC	9413 9402	0.8	61.2	< 0.05	< 0.05	7.4	0.04	0.26	1.1	17	0.3	4.1	30	24.5	10
FC-02-01-9.9M	9413 9402	2.0	46.2	0.60	< 0.05	16.8	0.21	0.70	1.8	65	1.3	8.2	32	94.5	< 10
FC-02-01-10.6M	9413 9402	1.0	307	0.15	< 0.05	8.0	0.08	0.34	1.7	34	0.6	9.8	92	35.5	< 10
FC-02-01-23.1M	9413 9402	31.6	36.8	0.50	< 0.05	17.8	0.21	0.86	2.5	71	3.0	8.2	74	79.0	< 10
FC-02-01-66.4M	9413 9402	1.6	77.6	0.20	< 0.05	15.0	0.12	0.36	2.3	53	0.6	8.2	68	91.5	< 10
FC-02-01-92.2M	9413 9402	1.2	70.6	0.15	< 0.05	13.6	0.10	0.42	1.3	41	0.6	5.6	48	60.5	< 10
FC-02-03-44.6M	9413 9402	1.0	968	0.40	< 0.05	8.2	0.41	0.18	5.9	112	1.1	15.7	86	136.0	20
FC-02-03-60.3M	9413 9402	0.8	255	0.40	< 0.05	2.6	0.70	0.18	1.3	237	1.5	7.8	106	31.5	< 10
FC-02-03-73.5M	9413 9402	2.8	126.5	0.70	< 0.05	25.8	0.29	1.10	4.7	88	2.3	10.8	78	102.0	< 10
FC-02-03-83.7M	9413 9402	0.4	282	0.30	< 0.05	2.2	0.37	0.06	0.3	217	0.4	4.9	96	24.0	< 10
FC-02-03-89.7M	9413 9402	1.2	101.0	0.20	< 0.05	13.6	0.11	0.34	2.1	44	0.7	6.1	64	65.0	< 10
FC-02-05-15.3M	9413 9402	10.8	22.6	0.05	< 0.05	7.6	0.13	0.44	4.8	111	3.0	8.4	26	19.5	< 10
FC-02-05-27.5M	9413 9402	16.8	61.3	0.20	< 0.05	9.8	0.07	0.20	1.4	37	4.1	4.3	292	41.5	< 10
FC-02-05-43.8M	9413 9402	14.0	68.4	0.30	< 0.05	13.4	0.09	0.26	1.8	42	3.2	5.4	276	71.5	10
FC-02-05-60.8M	9413 9402	43.4	55.6	0.30	< 0.05	15.4	0.13	0.90	3.0	63	3.1	5.6	130	68.0	< 10
FC-02-05-77.1M	9413 9402	4.4	86.5	0.15	< 0.05	15.4	0.09	0.46	1.8	23	1.2	7.6	32	61.5	< 10
FC-02-05-89.65M	9413 9402	2.0	218	0.05	< 0.05	13.0	0.06	0.74	1.1	56	0.5	10.1	48	47.5	< 10
FC-02-05-117.5M	9413 9402	1.0	25.0	0.05	< 0.05	10.4	0.06	0.62	1.5	17	1.0	3.9	50	32.0	< 10
FC-02-05-137.7M	9413 9402	3.4	39.0	0.25	< 0.05	21.2	0.12	2.62	3.2	40	1.8	8.2	68	101.0	< 10
FC-02-05-153.3M	9413 9402	17.2	134.0	0.30	< 0.05	20.0	0.21	0.38	2.2	83	1.7	9.1	86	>500	< 10

CERTIFICATION: [signature]

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Account : PFD

CERTIFICATE OF ANALYSIS A0221518

SAMPLE	PREP CODE	Weight Kg	Au ppb ICP	Pt ppb ICP	Pd ppb ICP	Ag ppm (ICP)	Al % (ICP)	As ppm	Ba ppm (ICP)	Be ppm (ICP)	Bi ppm (ICP)	Ca % (ICP)	Cd ppm (ICP)	Ce ppm (ICP)	Co ppm (ICP)	Cr ppm (ICP)	Cs ppm (ICP)	Cu ppm (ICP)	Fe % (ICP)	Ga ppm (ICP)
FC-02-06-26.6M	94139402	0.82	2	< 5	< 2	0.08	5.90	43.0	391.0	0.95	0.40	1.85	0.10	71.2	23.0	113	1.60	15.6	4.93	14.80
FC-02-06-42.7M	94139402	0.60	< 2	< 5	< 2	0.26	4.70	10.6	491.5	0.85	0.46	0.06	0.42	68.6	7.9	66	1.55	23.0	4.02	11.35
FC-02-06-61.1M	94139402	0.50	4	< 5	4	0.42	7.82	238	382.0	0.85	0.61	1.75	0.52	61.3	70.7	290	1.00	145.7	9.81	21.95
FC-02-06-77.8M	94139402	0.82	< 2	< 5	< 2	0.22	5.02	12.6	542.9	1.55	0.45	4.30	0.16	65.6	7.2	55	1.60	18.6	2.90	11.30
FC-02-06-101.4M	94139402	0.80	< 2	< 5	< 2	0.06	6.88	38.0	466.0	1.80	0.12	0.72	0.12	81.3	11.3	80	2.20	4.8	2.75	18.85
FC-02-06-115.0M	94139402	0.90	< 2	< 5	< 2	0.20	4.00	5.0	265.5	1.05	0.52	0.39	0.10	46.4	3.7	109	1.30	11.0	1.28	8.90
FC-02-06-132.9M	94139402	0.68	< 2	< 5	< 2	0.08	5.62	19.4	449.5	1.25	0.25	0.13	0.08	72.9	15.1	79	1.90	45.4	4.40	15.45
ACE-02-03-119.0M	94139402	0.58	< 2	5	< 2	0.08	9.77	1.8	815.2	3.50	0.20	0.42	0.04	93.6	24.3	94	3.65	33.6	4.95	26.95
ACE-02-03-122.0M	94139402	0.86	< 2	< 5	< 2	0.08	11.05	1.2	845.2	3.45	0.21	0.37	0.04	122.5	27.7	88	3.50	60.5	5.84	30.15
ACE-02-04-17.1M	94139402	1.04	28	< 5	< 2	0.06	7.43	0.6	1281.0	2.50	0.10	1.15	0.02	110.5	15.2	83	4.85	23.0	2.99	20.00
ACE-02-04-28.0M	94139402	0.74	< 2	< 5	< 2	0.08	7.78	0.4	1135.5	2.55	0.08	0.55	0.02	99.2	19.3	68	2.35	23.2	4.17	21.85
ACE-02-04-37.3M	94139402	0.46	< 2	< 5	< 2	0.14	7.60	3.4	125.0	1.90	0.52	4.10	1.16	72.6	16.3	81	3.25	16.6	3.28	16.45
ACE-02-04-50.0M	94139402	0.44	6	< 5	< 2	0.08	8.31	28.0	35.0	3.35	0.79	1.95	0.56	70.2	16.3	90	2.15	37.6	3.96	23.40
ACE-02-04-71.3M	94139402	0.90	< 2	< 5	< 2	0.10	9.30	0.2	426.0	3.20	0.13	4.80	0.26	75.9	23.1	74	10.60	55.8	3.53	25.40
ACE-02-04-83.0M	94139402	0.60	< 2	< 5	6	0.44	5.25	1.8	120.5	1.65	1.69	2.70	1.58	43.6	10.4	101	2.80	28.0	2.55	13.60
ACE-02-04-98.1M	94139402	1.04	< 2	< 5	< 2	0.08	9.74	0.8	1427.0	3.45	0.15	0.39	0.04	118.5	26.6	109	3.90	28.4	5.24	28.40
ACE-02-04-112.0M	94139402	0.86	< 2	45	< 2	0.14	10.65	0.6	1440.5	3.75	0.30	0.33	0.02	120.5	31.2	87	4.25	53.6	6.20	29.40

CERTIFICATION: [illegible signature]

ALS Chemex
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PHONE: 604-984-0221 FAX: 604-984-0218

To: BARKER MINERALS LTD.

22117 37A AVE.
LANGLEY, BC
V2Z 1N9

Project : TJB SERIES
Comments: ATTN: LOUIS E. DOYLE CC: TIM BARRETT

Certificate Date: 13-AUG-200
Invoice No. : I0221518
P.O. Number :
Account :PFD

CERTIFICATE OF ANALYSIS A0221518

SAMPLE	PREP CODE	Ge ppm (ICP)	Hf ppm	In ppm	K % (ICP)	La ppm (ICP)	Li ppm (ICP)	Mg % (ICP)	Mn ppm (ICP)	Mo ppm (ICP)	Na % (ICP)	Nb ppm (ICP)	Ni ppm (ICP)	P ppm (ICP)	Pb ppm (ICP)	Rb ppm (ICP)	Re ppm	S %	Sb ppm (ICP)	Se ppm
FC-02-06-26.6M	9413 9402	0.20	4.2	0.110	1.41	37.0	85.8	2.17	985	0.85	0.46	3.2	62.1	470	12.5	76.8	0.002	0.37	0.45	1
FC-02-06-42.7M	9413 9402	0.20	2.3	0.150	1.53	34.5	22.4	1.09	805	0.40	0.18	2.8	13.0	130	35.5	79.8	0.002	0.11	0.35	< 1
FC-02-06-61.1M	9413 9402	0.25	1.4	0.085	1.06	29.5	125.0	5.66	2150	1.45	0.29	21.2	210	1510	72.0	53.1	0.004	0.86	1.00	1
FC-02-06-77.8M	9413 9402	0.15	1.8	0.155	1.86	30.5	8.4	1.67	1045	0.40	0.32	2.7	15.6	110	14.5	87.0	0.002	0.05	0.15	< 1
FC-02-06-101.4M	9413 9402	0.20	2.7	0.040	2.27	42.5	20.0	1.02	490	0.45	0.56	3.4	26.6	410	14.5	112.5	0.004	0.10	0.25	< 1
FC-02-06-115.0M	9413 9402	0.15	1.3	0.020	1.29	22.5	9.4	0.50	185	0.80	0.33	1.8	9.6	160	18.5	59.8	0.002	0.01	0.10	< 1
FC-02-06-132.9M	9413 9402	0.20	2.3	0.140	1.77	38.0	43.4	1.58	675	0.50	0.13	2.7	31.4	330	9.0	92.4	0.002	0.34	0.35	< 1
ACE-02-03-119.0M	9413 9402	0.25	1.3	0.080	2.70	50.5	37.4	1.22	415	0.40	1.16	10.0	49.0	590	17.0	142.5	0.004	0.41	0.05	1
ACE-02-03-122.0M	9413 9402	0.30	16.8	0.095	2.64	66.0	40.6	1.23	395	2.45	1.18	12.6	47.4	810	13.5	138.0	0.004	0.67	0.10	1
ACE-02-04-17.1M	9413 9402	0.25	0.8	0.055	2.72	56.5	22.6	0.70	570	0.65	1.80	16.5	27.0	1120	8.5	117.0	0.004	0.21	0.05	1
ACE-02-04-28.0M	9413 9402	0.30	0.6	0.070	2.38	50.5	27.2	0.97	425	0.70	1.31	18.5	30.4	1090	8.0	107.0	0.004	0.26	< 0.05	1
ACE-02-04-37.3M	9413 9402	0.20	1.2	0.025	0.61	37.0	14.4	1.26	225	3.55	5.44	6.7	37.8	1330	51.5	36.7	0.006	2.48	0.20	3
ACE-02-04-50.0M	9413 9402	0.25	1.1	0.025	0.86	36.0	21.0	0.90	540	50.70	5.64	5.9	60.6	190	12.5	48.3	0.032	3.62	0.70	7
ACE-02-04-71.3M	9413 9402	0.25	1.4	0.055	3.00	47.5	48.6	1.64	380	1.10	3.26	5.3	46.2	460	18.0	142.5	0.004	1.29	0.10	2
ACE-02-04-83.0M	9413 9402	0.15	0.2	0.020	0.52	24.5	15.0	0.74	700	131.50	3.76	2.5	65.8	2700	59.0	34.3	0.122	1.75	0.05	3
ACE-02-04-98.1M	9413 9402	0.30	0.8	0.085	2.70	62.5	37.0	1.23	580	1.95	1.07	13.3	52.8	540	13.5	145.0	0.004	0.16	< 0.05	1
ACE-02-04-112.0M	9413 9402	0.30	0.6	0.090	2.94	64.5	39.2	1.34	450	0.95	1.04	13.2	57.1	450	27.5	159.5	0.004	0.27	0.05	1

CERTIFICATION: [signature]

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SAMPLE	PREP CODE		Sn ppm	Sr ppm (ICP)	Ta ppm (ICP)	Te ppm (ICP)	Th ppm (ICP)	Ti % (ICP)	Tl ppm (ICP)	U ppm (ICP)	V ppm (ICP)	W ppm (ICP)	Y ppm (ICP)	Zn ppm (ICP)	Zr ppm	Hg ppb
FC-02-06-26.6M	9413	9402	17.0	125.5	0.15	< 0.05	19.4	0.14	0.38	1.8	73	1.0	8.0	88	69.7	< 10
FC-02-06-42.7M	9413	9402	22.6	45.4	0.15	< 0.05	18.0	0.09	0.40	1.8	31	1.4	5.3	196	51.7	< 10
FC-02-06-61.1M	9413	9402	9.8	104.5	1.15	< 0.05	5.0	0.86	0.32	1.0	277	14.3	8.4	482	53.8	10
FC-02-06-77.8M	9413	9402	5.2	147.0	0.15	< 0.05	14.0	0.09	0.50	1.7	31	1.1	8.6	62	47.3	< 10
FC-02-06-101.4M	9413	9402	1.8	86.6	0.20	< 0.05	18.4	0.11	0.70	3.3	52	1.3	9.4	78	73.2	< 10
FC-02-06-115.0M	9413	9402	1.2	54.7	0.05	< 0.05	11.0	0.06	0.48	1.8	22	0.6	4.2	34	43.0	< 10
FC-02-06-132.9M	9413	9402	10.6	22.8	0.15	< 0.05	16.2	0.09	2.44	4.2	40	2.3	6.9	124	76.5	< 10
ACE-02-03-119.0M	9413	9402	2.6	166.5	0.45	< 0.05	16.8	0.18	0.64	1.8	89	0.8	8.1	110	34.9	< 10
ACE-02-03-122.0M	9413	9402	2.4	154.0	0.60	0.05	19.4	0.25	0.60	2.8	126	3.9	9.2	122	34.5	< 10
ACE-02-04-17.1M	9413	9402	2.0	147.0	0.75	< 0.05	22.2	0.58	0.48	3.0	80	1.6	25.2	54	22.6	< 10
ACE-02-04-28.0M	9413	9402	2.2	134.0	0.85	< 0.05	15.8	0.62	0.46	2.2	90	0.9	18.6	102	16.3	< 10
ACE-02-04-37.3M	9413	9402	0.6	463	0.30	0.15	14.6	0.19	0.34	4.6	152	0.8	11.5	360	44.0	< 10
ACE-02-04-50.0M	9413	9402	1.8	418	0.30	0.05	8.0	0.17	0.32	8.8	391	1.9	10.9	68	29.1	< 10
ACE-02-04-71.3M	9413	9402	2.4	428	0.25	0.05	14.6	0.21	0.86	3.8	126	1.0	9.7	58	40.2	< 10
ACE-02-04-83.0M	9413	9402	0.8	262	0.05	0.05	8.4	0.08	0.24	6.5	128	0.7	17.5	162	19.9	< 10
ACE-02-04-98.1M	9413	9402	2.8	141.0	0.60	< 0.05	19.6	0.26	0.70	2.2	101	0.6	20.7	136	23.6	< 10
ACE-02-04-112.0M	9413	9402	2.6	142.5	0.60	< 0.05	19.4	0.26	0.70	2.3	113	0.4	20.4	146	18.2	< 10

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Page Number : 1
Total Pages : 1
Certificate Date: 20-AUG-2002
Invoice No. : I0222088
P.O. Number :
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CERTIFICATE OF ANALYSIS A0222088

SAMPLE	PREP CODE		Pb %	Zn %
02115	212	--	6.40	17.30

CERTIFICATION: C. [signature]

OVERLIMITS from A0221518

Preliminary Report

on

Diamond Drilling and Trenching

for the

Frank Creek & Ace Projects

Cariboo Mining Division

N.T.S. 93A/11, 93A/14

Owner/Operator:
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June 26, 2002

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1.0 Summary

A total of 1,459.1 metres of diamond drilling was completed on 11 holes split between the Frank Creek and Ace Project areas, located approximately 20 to 40 kilometres northeast of Likely, B.C. The property is 100% owned by Barker Minerals Ltd. Table 1 shows the breakdown of drilling between the project areas and individual holes, as well as hole coordinates and orientations.

Table 1 2002 Diamond Drillholes

Hole No.	Date Collared	Date Completed	Northing	Easting	Elevation	Azimuth	Dip	Length
			metres	metres	metres			metres
FC-02-01	15-May-02	16-May-02	57 + 95 N	16 + 00 W	1042	090	-45	97.84
FC-02-02	16-May-02	18-May-02	57 + 20 N	12 + 80 W	1077	090	-45	144.78
FC-02-03	18-May-02	20-May-02	54 + 30 N	13 + 47 W	1140	090	-45	118.87
FC-02-04	20-May-02	22-May-02	54 + 50 N	12 + 20 W	1140	090	-45	158.19
FC-02-05	22-May-02	24-May-02	55 + 60 N	15 + 00 W	1115	075	-45	158.50
FC-02-06	24-May-02	25-May-02	56 + 11 N	15 + 41 W	1103	070	-45	135.03
						Total-FC		**813.21**
Ace-02-01	26-May-02	28-May-02	16 + 15S	6 + 65E	1005	180	-45	144.78
Ace-02-02	29-May-02	31-May-02	18 + 80S	8 + 64E	1025	180	-45	164.90
Ace-02-03	31-May-02	2-Jun-02	4 + 70N	5 + 70W	1028	180	-45	150.88
Ace-02-04	2-Jun-02	3-Jun-02	4 + 20N	5 +00W	1025	180	-45	117.35
Ace-02-05	3-Jun-02	4-Jun-02	5 + 00N	6 + 25W	1027	n/a	-90	67.97
						Total-Ace		**645.87**
						Total		**1,459.08**

On Frank Creek, all holes were located within the F-1 Target Area. Four of the 6 holes tested the Frank Creek mineralized horizon, the other two tested the immediate structural footwall, to a wide, conductive graphitic fault zone. FC-02-01, 05, and 06 intersected disseminated to semi-massive sulphides, dominantly pyritic, near the projected location of the mineralized horizon, downslope (northwest) of the original Frank Creek (F-1) Showing. FC-02-03 tested under the showing and intersected approximately 50cm of pyritic massive sulphide similar to that in the showing, caught in a large gougy fault zone. FC-02-02 and 04 tested strong max-min conductors and intersected significant graphitic shear zones in carbonate-altered metasedimentary rocks.

At the Ace, ACE-02-01 tested the 16S Target, defined by a max-min conductor and a subtle gravity anomaly. ACE-02-02 tested a ground magnetic high flanking the 16S anomalies to the south. The first hole intersected approximately 6.5 metres of "felsite", a fine-grained plagioclase-rich unit that elsewhere hosts massive sulphide mineralization.

Three holes, ACE-02-03 to 05, tested the 5N Zone, 2 kilometres to the west. All three intersected at least 40 metres of "felsite"; two intersected significant sulphide mineralization near the top of the interval. The three holes followed up on two holes completed through the felsite unit in 1998. Drilling also tested high ground magnetics, a max-min conductor, and another subtle gravity anomaly.

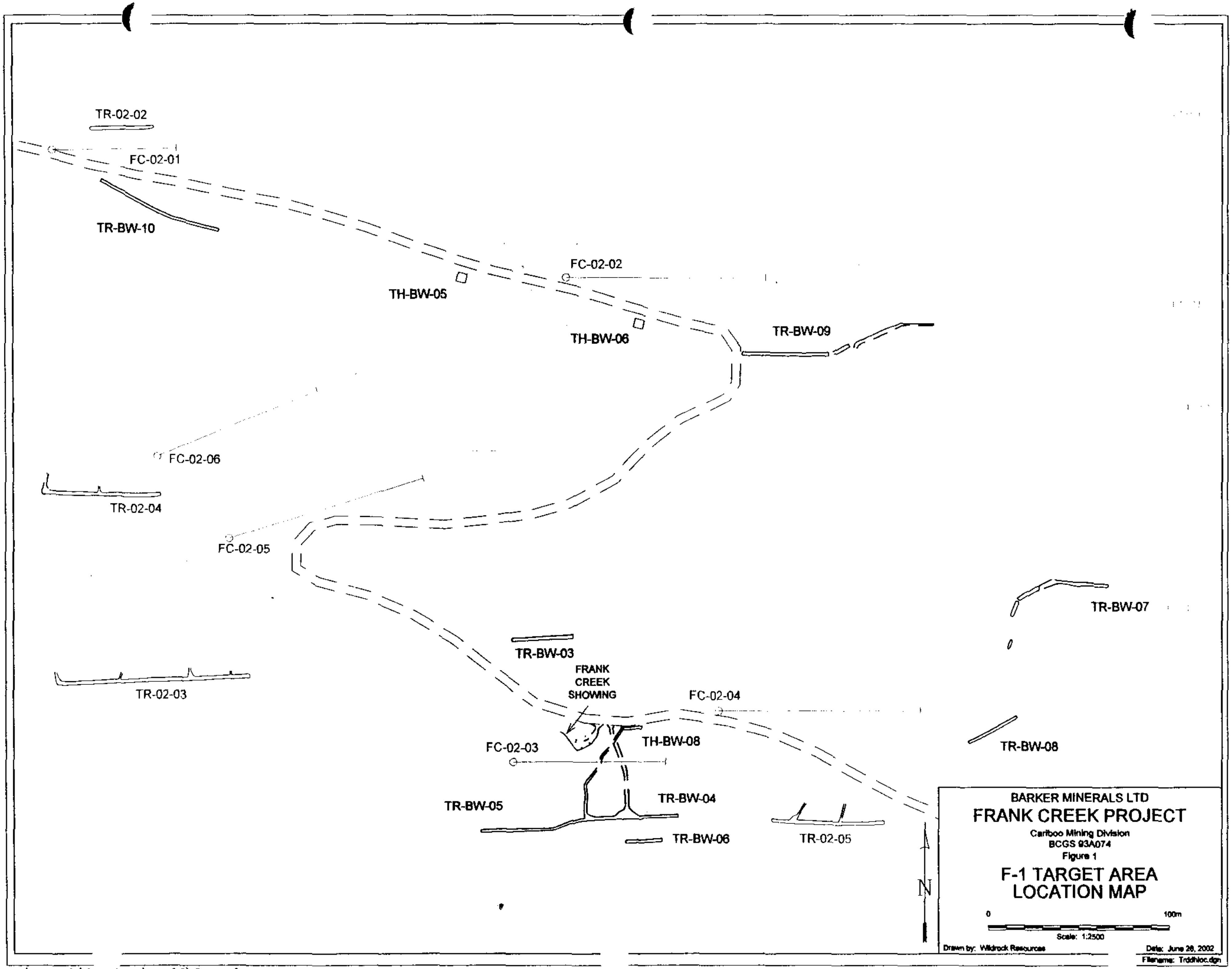
A total of 154 samples were collected from the core by sawing marked intervals, bagging half the core from the interval and retaining the other half as a permanent core record. A total of 84 samples were collected from the Ace core and 70 samples from Frank Creek. Sample lengths averaged 1.0-1.5 metres. Samples were transported to EcoTech Laboratories in Kamloops for analysis. All samples will be analyzed for 28 elements using standard ICP analysis after aqua regia leach. Au, Pt, and Pd will be fire assayed with a 30g sample. Trace elements Hg, Se, Te, Tl, Sn, and B will be analyzed using ICP methods at Acme Analytical Laboratories in Vancouver. A total of 71 samples will also be run for whole rock major elements at EcoTech in an attempt to better characterize host rocks. Petrographic work is also planned for selected samples.

In addition to diamond drilling, a series of excavator trenches tested several targets in and adjacent to the F-1 Target Area at Frank Creek. Five trenches were excavated over a total distance of 289 metres; details are shown in Table 2. Exposures in trenches 1, 3, 4 and 5 provided much structural data and will be invaluable in piecing together the geology of the area. All but TR-02-05 had been reclaimed at the time of writing.

Table 2 2002 Trench Log

Trench	Length (metres)	Location & Target	Description
TR-02-01	20	Northwest flank of magnetic high.	Argillite, dark phyllites.
TR-02-02	36	NW extension of mineralized horizon.	Qtz-ser phyllite, argillite, unstable.
TR-02-03	107	Max-min conductor, magnetic high.	Siliceous, gritty phyllites.
TR-02-04	65	Max-min conductor, magnetic high.	Argillite, minor sericitic phyllite.
TR-02-05	61	IP anomaly, near ms boulders.	Siliceous grits, minor argillite.

Figure 1 shows the locations of trenches and drill collars on the F-1 Target area. Trench TR-02-01 is located immediately west of F-1, off the map. Figures 2 and 3 show the drillhole locations on the 16S Target and 5N Zone on the Ace, respectively.



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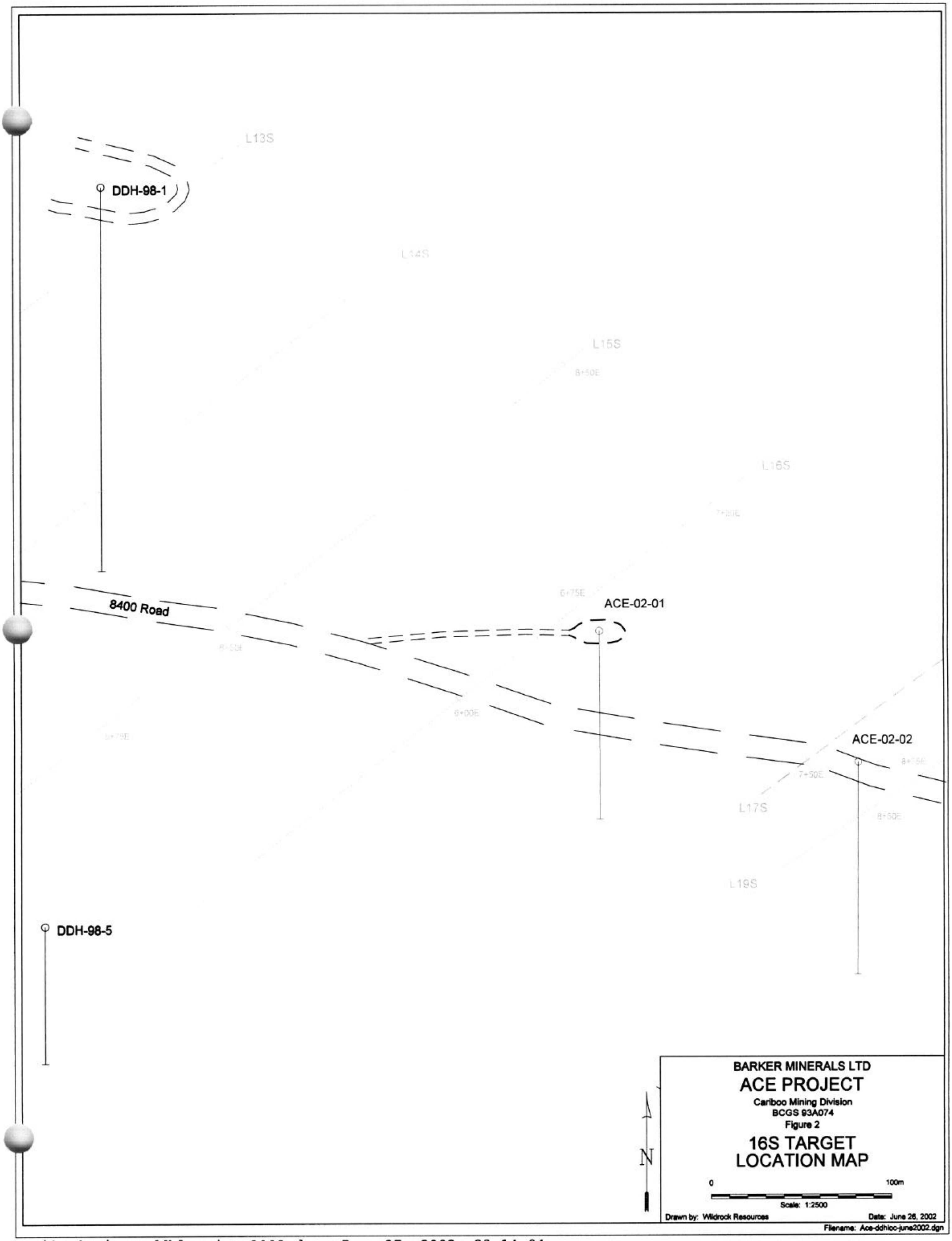

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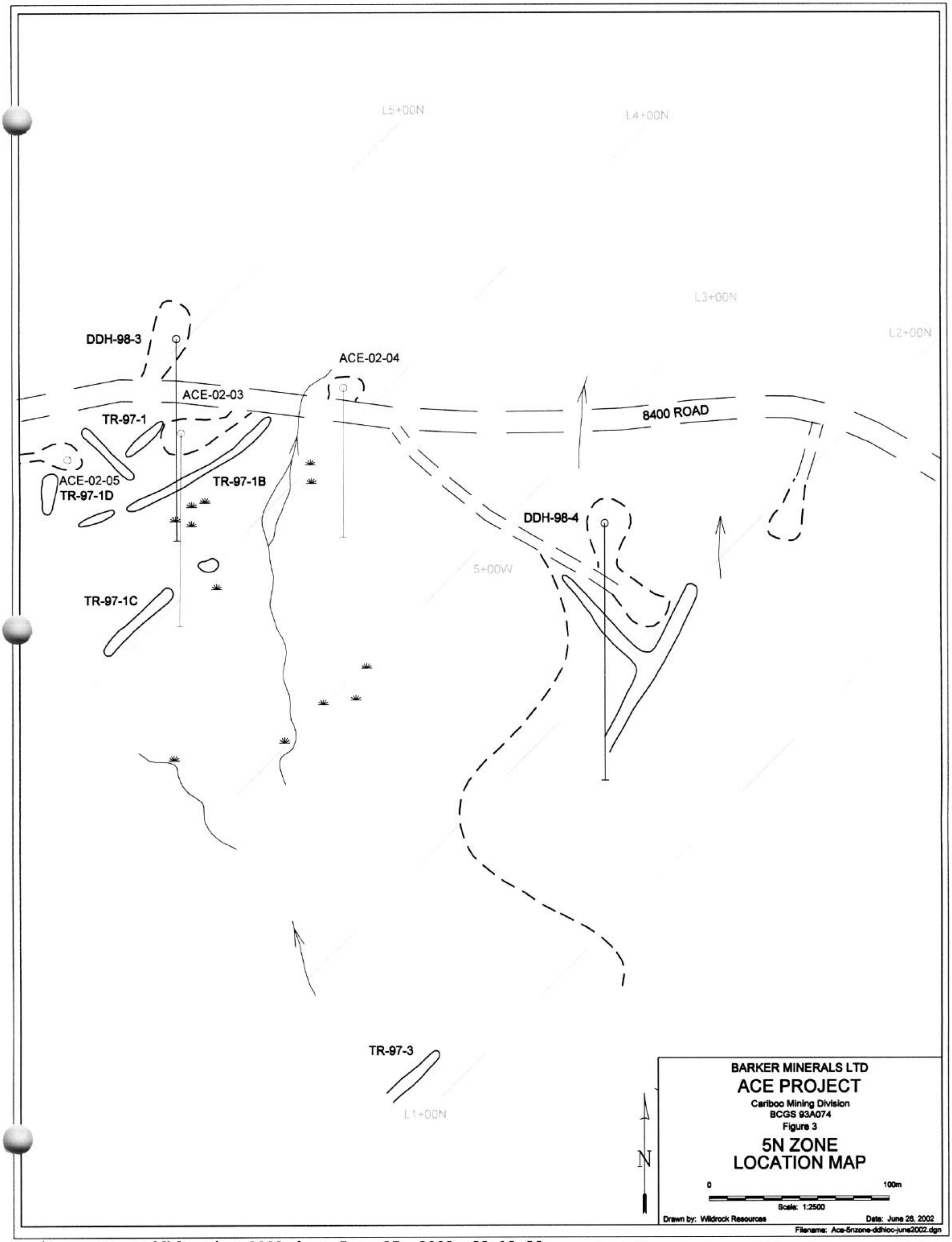

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2.0 Diamond Drilling and Trenching

2.1 *Frank Creek Trenching*

TR-02-01

A trench 100 metres in length was originally planned across a magnetic high to low on line 54N, between 19+50W and 20+50W, approximately 750 metres west of the F-1 Showing. However, bedrock was not encountered in test pits along the planned length. Bedrock was exposed 30 metres to the west, consisting of dark chloritic phyllite (QCSP) or argillite, quartz-sericite phyllite (QSP), and a siliceous and weakly pyritic exhalative unit, less than 50cm thick. Strikes are northwesterly with intermediate to steep southwesterly dips. A well-developed crenulation cleavage indicates proximal third phase folding.

TR-02-02

A second trench was excavated on line 58N, between 15+45W and 15+80W. The trench tested for sulphide mineralization approximately 40 metres north of mineralization uncovered in TR-BW-10 and adjacent to drillhole FC-02-01. High subsurface water flow made the trench walls highly unstable and only muck piles were investigated. Rock consists of waxy quartz-sericite phyllite and less dark chloritic phyllite or argillite. No significant mineralization was encountered.

TR-02-03

Good exposure was found in a 107 metre long trench immediately south of line 55W, between 15+00W and 16+00W. The centre of the trench is 225 metres west of the F-1 Showing. The trench cuts a max-min conductor in an area of strong magnetic gradient. Most of the trench is underlain by grit and siliceous phyllites (QSP and QCSP). No obvious conductors or moderately magnetic rocks were mapped. Structurally, the principal foliation strikes to the north with intermediate west dips at the east end of the trench. Toward the middle of the trench, the foliation is folded to the west-northwest and back more northwesterly further west. No significant mineralization was encountered.

TR-02-04

TR-02-04 was excavated 100 metres north of TR-02-03, also on a max-min conductor. Much of the trench is underlain by variably graphitic dark phyllite or argillite. Structurally, the argillite is strongly rodded with a west-northwesterly strike on the foliation. A gougy fault separates the argillite from softer, more sericitic phyllites at the west end of the trench. Strikes turn dramatically to the north-northeast. Again, no significant mineralization was encountered in the trench.

TR-02-05

The last trench of the spring program is located 50 metres east of TR-BW-04, 100-150 metres southeast of the Frank Creek Showing (Figure 1). The trench is underlain by interlayered grit and argillite, folded into rods and fold crests that plunge gently to the northwest. The main foliation strikes to the northwest and dips steeply to the southwest. No significant mineralization was encountered.

2.2 Frank Creek Diamond Drillholes

FC-02-01

The first hole, FC-02-01, was collared on line 58N at 16+00W, to test for mineralization uncovered near the west end of TR-BW-10 (Figure 1). Blebby pyrite with 2% patchy cp and minor sphalerite and galena was encountered in two narrow sections around 27.0 and 29.2 metres downhole. Host rocks are grey quartz sericite phyllites (QSP), dark chloritic phyllites (QCSP) and argillite. Gougy and brecciated faults are common throughout. Recoveries were generally >90% overall, lower around these faults. Wispy tan carbonate and spotted ankerite alteration are relatively common. Twelve samples were split for trace element analysis, 4 of those will also be analyzed for major elements.

FC-02-02

The second hole was collared just north of line 57N, near 12+80W, and drilled under the first switchback and TR-BW-09. The hole encountered quartz-sericite phyllite, often gritty and flooded with wispy tan carbonate and associated Cr-mica. The hole also intersected several large graphitic fault zones with associated deformation breccias. These breccias frequently consist of QSP clasts in a dark, weakly carbonaceous matrix. Clasts are attenuated along the principal foliation, presumably to the northwest. The hole was stopped short of the inferred contact with limestones mapped over the east half of TR-BW-09. A total of 7 samples were split, 4 of which will be analyzed for major elements.

FC-02-03

The third hole was collared around 35 metres west of the F-1 Showing and drilled to the east under the showing. The top of the hole is dominated by a 17 metre thick section of fault gouge (true thickness ~12 metres). Within the fault zone, a couple of pieces of pyritic massive sulphide were recovered over an interval around 50cm long. The Frank Creek massive sulphide has apparently been down dropped to the west, as indicated by the position of these sulphide pieces in the centre of the fault, below the showing. The footwall consists of quartz-sericite (QSP) and quartz-chlorite phyllites (QCSP) interrupted by numerous gougy faults. As in FC-02-02, carbonate alteration is widespread. A total of 6 samples were split, 4 of which will be analyzed for major elements.

FC-02-04

The fourth hole was collared 135 metres east of FC-02-03, as part of a fence across the showing area. A line of IP run down the road across the showing area shows significant chargeabilities and low resistivities 100 metres east of the showing. Nothing that easily explains the anomaly was encountered; much of the section consists of QSP and QCSP, punctuated with thin argillite and limestone breccia intervals. A thick (13.5m) gougy quartz-graphite fault, likely part of Conductor B, dominates the lower third of the hole. Mineralization is quite weak and again, only 6 samples were split from this hole.

FC-02-05

FC-02-05 was collared on line 15W, at 55+60N, and drilled at –45° toward 75°. The hole was positioned to test strong soil geochemical anomalies and potential mineralization associated with a highly altered outcrop on D Road. The top of the hole consists of coarsely interlayered QSP and argillite. Disseminated mineralization including pyrite and minor sphalerite, chalcopyrite and galena, begins near the top of the hole, culminating in 1.3 metres of semi-massive to massive sulphides. Sulphides are hosts in quartz-sericite phyllite, possibly a felsic volcanic unit. Disseminated mineralization is far less extensive below the intersection, becoming grittier with significant deformation breccia toward a fault near the bottom of the hole. Unlike the other holes, faulting was weak through most of the hole. A total of 20 samples were split for analysis.

FC-02-06

In response to the discovery of significant mineralization in FC-02-05, a sixth hole was collared around 60 metres to the northwest, along strike, and drilled at –45° toward 70°. As expected, the hole is very similar to FC-02-05. The top of the hole consists of argillite and QSP with mineralization beginning near the top of a thick, relatively weakly faulted section of QSP. Mineralization consists of disseminated pyrite, minor chalcopyrite, sphalerite and traces of galena in layers and stringer zones. The sulphide horizon is marked by 10-12cm of semi-massive sulphides, similar to but far thinner than in FC-02-05. The footwall section consists of a carbonate-altered fine to medium-grained QSP, with local sections of deformation breccia and grit with very little faulting. A total of 19 samples were split for analysis.

2.3 Ace Diamond Drillholes

ACE-02-01

Two diamond drillholes; ACE-02-01 and 02, were collared in the 16S Zone just to the west of a two-drillhole fence (98-01 and 05) completed in 1998 (Figure 2). The first hole, ACE-02-01, tested a coincident max-min conductor and modest gravity anomaly on L16S near the 8400 Road. The top 49.0 metres consist of very siliceous quartz-muscovite schist and argillite. The next 10 metres consist of strongly calcareous marble, calc-schist, and argillite, immediately above a 6.5 metre section of "felsite". Felsite consists of plagioclase or albite with minor micas and quartz and elsewhere is the host for massive to disseminated mineralization (Payne, 1998). A hole drilled 190 metres to the west, 98-01, intersected 3.5 metres of felsite. The footwall is made up of quartz-rich, muscovite and chlorite schists with variable amounts of garnet. A total of 22 samples were collected from ACE-02-01, to better characterize the various units in the Ace stratigraphic section.

ACE-02-02

The second hole into the 16S Zone, ACE-02-02, was collared 100 metres to the east of ACE-02-01 on the 8400 Road (Figure 2), and tested a ground magnetic high 150 metres south of 8400 Road. The hole, ACE-02-02, did not intersect any felsite but did encounter two significant marble sections. The rest of the hole is made up of quartz-muscovite-chlorite+/-garnet schist. Although ACE-02-02 is situated near ACE-02-01, there is no obvious correlation of stratigraphy between the two. Seven samples were collected from throughout the hole.

ACE-02-03

Three drillholes, ACE-02-03 to 05, tested the 5N Zone, an area tested by extensive trenching and 2 diamond drillholes completed in 1998 (Figure 3). Drillhole 98-03 intersected 0.25 metres and 0.2 metres of massive sulphide near the top an 81.5 metre thickness of felsite. ACE-02-03 was collared approximately 52 metres south of 98-03 in semi-massive sulphide up-dip from the massive sulphides in 98-03. As in 98-03, weakly mineralized felsite extended an additional 69 metres below the strongly mineralized layer. The footwall consists of mainly weakly chloritic quartz-muscovite schist, argillite, and minor marble. Significantly more argillite is present in ACE-02-03. A total of 19 samples were split from ACE-02-03, 12 from the felsite.

ACE-02-04

ACE-02-04 was collared 93 metres east of, and between 98-03 and ACE-02-03 (Figure 3). Approximately 70.5 metres of felsite were encountered between 33.8 and 104.3 metres downhole, with disseminated and semi-massive mineralization spread over 4.9 metres between 53.8 and 58.7 metres downhole. The structural hangingwall consists of weakly biotitic and chloritic quartz-muscovite schists and a significant dirty marble. No such marble was identified in 98-03, but is present over 2 kilometres to the east, in ACE-02-01. The footwall consists of weakly chloritic quartz-muscovite schist. A total of 31 samples were split for analysis, 21 from the felsite unit.

ACE-02-05

The last hole, ACE-02-05, was collared on L5N at 6+25W and drilled vertically to test a coincident ground magnetic high and subtle gravity anomaly (Figure 3). Both surveys had suggested a southerly dip to the target and fold geometries seen a few hundred metres to the southeast supported the possibility that the target horizon could turn over to the south. ACE-02-05 was collared in the felsite unit, likely below the sulphide horizon. Here, the felsite fingers in and out with quartz-muscovite schist and argillite to a depth of 66.5 metres with felsite intervals becoming thinner and less frequent. Sulphide content is less than 10% combined pyrite and pyrrhotite, but is probably sufficient to explain both the high magnetic and gravity. Only 4 samples were collected from the felsite and one from a quartz vein.

3.0 Discussion

3.1 Frank Creek

A much clearer picture of the geology of the F-1 Target Area at Frank Creek has emerged from recently completed trenching and drilling. A thick package of siliceous, sericitic and weakly chloritic phyllites and quartz eye grits host the mineralized horizon, best exposed at the F-1 Showing. This section appears to be siliciclastic, although there is some potential for felsic volcanic and/or volcaniclastic rocks within the package. Quartz eye units have previously been interpreted as felsic volcanic rocks and more study is required to determine their true nature. Much of the section consists of dark, argillaceous phyllites interpreted to be mudstones, and pale grey, fine-grained to weakly gritty phyllites interpreted to be siltstones. Occasional limestone units add to the sedimentary character of the section.

Structurally, trenching has shown that the rocks are tightly folded around northwest-plunging folds in an apparent "S" pattern, suggesting a large closure may exist to the southwest. Earlier recumbent, likely isoclinal second phase folds appear to have similar northwest-plunging fold axes, but have been folded around the more obvious third phase features. Faulting consists of more plastic shear zones that are parallel to the principal foliation (S2), with associated deformation breccias. Clasts are mainly siliceous metasediments, attenuated along F2, supported in a fine-grained dark grey muddy matrix. Late steep, north-trending brittle block faults cut these earlier features.

Mineralization includes pyritic massive sulphides exposed in the F-1 discovery trench. Drilling has intersected significant strongly disseminated, semi-massive to massive pyrite-rich mineralization along trend to the northwest in FC-02-05, 06, and 01. This mineralized trend was also exposed in TR-BW-10. Besides pyrite, chalcopyrite, sphalerite, and galena were identified in relatively minor amounts.

Future work on the F-1 Target Area should focus on extending the mineralized trend to the northwest where it likely continues to near the location of massive sulphide boulders in placer workings in Frank Creek. Efforts should also be made to extend the trend to the southeast, upslope from the showing area. Given the structural and lithological complexity of the area, emphasis should be on trying to understand the geology through trenching and drilling. Work should also focus on soil geochemistry to provide high quality targets for this work. Finally, the F-1 Target Area will be used as a template for exploring the many other targets on the Frank Creek and SCR Project Areas.

3.2 Ace Project

Drilling and mapping in 1998 provided an excellent geological framework for the project area. Delineation of the "felsite" unit lead to the discovery of massive to semi-massive mineralization in the 5N Zone, an area that was extensively trenched prior to 1998. Three holes drilled into the zone this spring have defined the felsite package as a unit some 60-80 metres thick, possibly thickened by a series of tight to isoclinal folds. Immediately to the west in ACE-02-05 and the east in DDH-98-04, the unit breaks up into a series felsite layers, separated by weakly chloritic quartz-muscovite schists.

The felsite unit is reportedly made up of up to 80% plagioclase (Payne, 1998), although at least one sample was composed of mainly albite (Walus, 1997). The fine-grained nature of the rock, its intimate relationship to a series of thin marble bands, and almost regional extent suggests that it may be an exhalative horizon and an excellent target horizon for VMS deposits. Sulphide mineralization in 98-3, ACE-02-03, and ACE-02-04 is located near the structural top of the unit. Extensive whole rock analysis of the felsite in ACE-02-03 and 04 should help to characterize the unit.

In the 16S Area, ACE-02-01 was collared to test coincident max-min conductor and subtle gravity anomaly on line 16S. Both the conductor and gravity were caused by a 6.5-metre thickness of felsite with up to 10% pyrite and pyrrhotite. A second hole, ACE-02-02, was collared immediately above the felsite and tested the footwall toward a large ground magnetic high to the south. That anomaly was not adequately explained in the core.

At the current time, the 16S Target Area does not warrant further drilling. Further compilation work is required for the 5N Zone to determine if further drilling is required there. To the east, soil sampling, is warranted on some of the recently established grid on the east half of the project area. Also to the east, a coincident max-min conductor and subtle gravity anomaly on lines 3800S to 4700S should be followed up by extending the grid and possibly extending the gravity coverage to adjacent lines. Drilling should be planned on the most promising part of the anomaly.

Respectfully submitted,

Christopher J. Wild, P.Eng.
Consulting Geological Engineer

June 26, 2002

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Appendix 1
Certificate of Qualifications

I, Christopher J. Wild, do hereby certify that:

1) I am a consulting geological engineer currently residing at 307 Lexington Road, Comp 25, RR3, Lexington Heights Subdivision, Williams Lake, British Columbia, V2G 1M3.

   a) I am a graduate of the University of British Columbia, with a Bachelor of Applied Science (B.A.Sc.) in Geological Engineering, Mineral Exploration Option (1984).

   b) I have worked in mineral exploration and mine geology mainly in Canada on a full-time basis since 1985. I am former Chief Geologist at the Goldstream Mine, near Revelstoke, B.C. and Mount Polley Mine, near Williams Lake, B.C.

   c) I am a Registered Member of the Association of Professional Engineers and Geoscientists of the Province of British Columbia (1994), and a member of the Canadian Institute of Mining and Metallurgy (CIM).

2) I supervised the diamond drilling and trenching program described herein between May 14 – June 18, 2002, and am responsible for all aspects of this report.

3) I am not aware of any material fact or material change with respect to the contents of this report that is not reflected in this report, the omission to disclose which makes this report misleading.

4) I am independent of Barker Minerals Limited or any of the claims described in this report, as set out in section 1.5 of NI 43-101.

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Christopher J. Wild, P.Eng.
Consulting Geological Engineer

June 26, 2002

Appendix 2
Summary of Core Samples

Summary of Samples and Requested Analyses

Ace Project						
Drillhole	**Sample Numbers**		**No of Samples**		**Samples for Whole Rock Analysis**	**No.**
ACE-02-01	8151	8172	22		8151-54, 57, 64, 66, 68, 70-72	11
ACE-02-02	8173	8179	7		8173, 75, 76, 79	4
ACE-02-03	8180	8198	19		8182, 83, 85, 87, 89-92, 95, 97, 98	11
ACE-02-04	8199	8229	31		8199-8206, 08, 10-14, 16, 18-26, 28, 29	18
ACE-02-05	8230	8234	5		8230, 34	2
		Total	**84**		**Total**	**46**
Frank Creek Project						
Drillhole	**Sample Numbers**		**No of Samples**		**Samples for Whole Rock Analysis**	**No.**
FC-02-01	8235	8246	12		8236, 38, 42, 46	4
FC-02-02	8247	8253	7		8247, 50, 52, 53	4
FC-02-03	8254	8259	6		8254, 56, 58, 59	4
FC-02-04	8260	8265	6		8260, 62, 65	3
FC-02-05	8266	8285	20		8271, 74, 81-85	7
FC-02-06	8286	8304	19		8286, 89, 92, 94, 98, 99, 8304	7
		Total	**70**		**Total**	**25**
All samples to be analyzed for 28-element ICP.						
All samples to be analyzed for Hg, Se Te.						
All samples to be analyzed for Au, Pt, Pd.						
Results (hard copy & email) and invoices to:						
Barker Minerals Ltd.						
22117 37A Ave.						
Langley, B.C.						
V2Z 1N9						
Attn: Louis Doyle, President						
email: barker@telus.net						
Tel: 604 530-8752						
Fax: 604 530-8751						
Results (hard copy & email), copy of invoices to:						
Wildrock Resources Consulting & Drafting						
Comp 25, RR3, Lexington Subdivision,						
Williams Lake, B.C.						
V2G 1M3						
Attn: Chris Wild, P.Eng.						
email: wildrock@telus.net						
Tel: 250 392-7688						
Fax: 250 392-7651						

2/28/03

Ace-02-01 Samples

ACE-02-01 Sample No.	From	To	Length	Lithology	ICP	WR
8151	**13.00**	**14.50**	**1.50**	**QMBS**	**x**	**x**
A0201-13	13.65	13.80	0.15			
A0201-47	16.75	16.8	0.05			
8152	**17.40**	**19.00**	**1.60**	**QMCS**	**x**	**x**
8153	**29.00**	**30.50**	**1.50**	**QCMS**	**x**	**x**
8154	**37.50**	**39.00**	**1.50**	**Arg**	**x**	**x**
A0201-12	41.15	41.30	0.15			
8155	**45.00**	**46.50**	**1.50**	**QMCS**	**x**	
8156	**47.50**	**49.00**	**1.50**	**Arg**	**x**	
8157	**49.00**	**49.70**	**0.70**	**CS**	**x**	**x**
8158	**49.70**	**50.70**	**1.00**	**Marb**	**x**	
8159	**50.70**	**52.10**	**1.40**	**Arg**	**x**	
8160	**52.10**	**54.00**	**1.90**	**Marb**	**x**	
8161	**54.00**	**55.20**	**1.20**	**QMCS**	**x**	
8162	**55.20**	**57.00**	**1.80**	**Arg**	**x**	
8163	**57.00**	**58.80**	**1.80**	**Arg**	**x**	
8164	**58.80**	**61.00**	**2.20**	**Fels**	**x**	**x**
8165	**61.00**	**63.00**	**2.00**	**Fels**	**x**	
8166	**63.00**	**65.30**	**2.30**	**Fels**	**x**	**x**
8167	**65.30**	**67.50**	**2.20**	**QCMS**	**x**	
A0201-14	65.40	65.60	0.20			
A0201-15	66.10	66.30	0.20			
A0201-04	71.56	71.63	0.07			
8168	**76.00**	**77.50**	**1.50**	**QMCGS**	**x**	**x**
8169	**88.00**	**89.50**	**1.50**	**QMCGS**	**x**	
8170	**103.60**	**105.10**	**1.50**	**QCMS**	**x**	**x**
8171	**130.00**	**131.50**	**1.50**	**QMCGS**	**x**	**x**
8172	**139.50**	**141.00**	**1.50**	**QMCS**	**x**	**x**
28	samples					

Ace-02-02 Samples

ACE-02-02 Sample No.	From	To	Length	Lithology	ICP	WR
8173	26.0	27.5	1.5	QMCGS	x	x
8174	47.9	48.1	0.2	QV	x	
8175	50.0	51.5	1.5	QMCS	x	x
8176	78.5	80.0	1.5	QMCGS	x	x
8177	110.0	111.5	1.5	QMCS	x	
8178	131.8	132.5	0.7	QV	x	
8179	151.0	152.5	1.5	QMCS	x	x
7	samples					

Ace-02-03 Samples

ACE-02-03 Sample No.	From	To	Length	Lithology	ICP	WR
8180	**8.50**	**9.70**	**1.20**	**SMS**	**x**	
8181	**9.70**	**11.80**	**2.10**	**DS**	**x**	
8182	**11.80**	**13.90**	**2.10**	**Fels**	**x**	**x**
8183	**13.90**	**16.00**	**2.10**	**Fels**	**x**	**x**
8184	**16.00**	**18.00**	**2.00**	**Fels**	**x**	
8185	**18.00**	**19.60**	**1.60**	**Fels**	**x**	**x**
8186	**19.60**	**21.00**	**1.40**	**Fels**	**x**	
A0203-16	21.00	21.60	0.60			
A0203-17	21.60	21.80	0.20			
A0203-18	21.80	22.35	0.55			
A0203-19	22.35	23.35	1.00			
A0203-20	23.35	23.95	0.60			
A0203-21	23.95	24.80	0.85			
A0203-22	24.80	25.70	0.90			
A0203-22A	25.90	26.90	1.00			
8187	**26.90**	**28.30**	**1.40**	**Fels**	**x**	**x**
8188	**28.30**	**30.00**	**1.70**	**Fels**	**x**	
A0203-23	30.40	31.10	0.70			
A0203-24	31.10	31.90	0.80			
A0203-25	31.90	32.75	0.85			
A0203-26	32.75	33.85	1.10			
A0203-27	33.85	34.45	0.60			
A0203-28	34.45	35.35	0.90			
A0203-29	35.35	35.80	0.45			
A0203-30	36.65	37.35	0.70			
A0203-31	38.50	39.40	0.90			
A0203-32	39.70	40.00	0.30			
8189	**40.00**	**41.50**	**1.50**	**Fels**	**x**	**x**
A0203-33	42.10	42.95	0.85			
A0203-34	42.95	43.45	0.50			
A0203-35	43.45	44.85	1.40			
A0203-36	44.85	45.65	0.80			
A0203-37	46.10	47.10	1.00			
A0203-38	47.10	48.20	1.10			
A0203-39	48.20	49.30	1.10			
A0203-40	49.30	50.30	1.00			
8190	**50.30**	**51.30**	**1.00**	**QMCS**	**x**	**x**
A0203-41	51.30	52.40	1.10			
A0203-42	52.40	53.00	0.60			
A0203-43	53.00	53.80	0.80			
A0203-44	53.80	54.25	0.45			
A0203-45	54.25	54.95	0.70			
A0203-46	54.95	55.50	0.55			
8191	**55.50**	**57.70**	**2.20**	**Arg**	**x**	**x**
8192	**57.70**	**59.60**	**1.90**	**Fels**	**x**	**x**
A0203-55	61.50	62.00	0.50			
8193	**64.00**	**65.50**	**1.50**	**Fels**	**x**	
8194	**69.30**	**70.80**	**1.50**	**Fels**	**x**	
8195	**72.60**	**74.60**	**2.00**	**Fels**	**x**	**x**

Ace-02-03 Samples

A0203-56	74.55	75.05	0.50			
A0203-57	75.25	76.40	1.15			
A0203-58	76.40	77.55	1.15			
A0203-54	80.00	81.25	1.25			
A0203-53	81.25	81.55	0.30			
8196	**95.10**	**97.00**	**1.90**	**QMCS**	**x**	
A0203-59	97.85	98.70	0.85			
A0203-60	97.90	98.45	0.55			
A0203-61	99.70	100.60	0.90			
A0203-62	100.60	101.45	0.85			
A0203-63	101.45	102.65	1.20			
A0203-64	104.20	104.90	0.70			
A0203-50	108.00	109.50	1.50			
A0203-49	109.30	109.50	0.20			
A0203-51	111.25	112.45	1.20			
A0203-52	112.45	113.20	0.75			
8197	**118.00**	**119.50**	**1.50**	**Arg**	**x**	**x**
8198	**132.00**	**133.50**	**1.50**	**QMCS**	**x**	**x**
A0203-65	138.38	138.68	0.30			
A0203-66	143.80	144.05	0.25			
69	samples					

Ace-02-04 Samples

ACE-02-04 Sample No.	From	To	Length	Lithology	ICP	WR
8199	17.50	19.00	1.50	QMBS	x	x
8200	25.50	27.00	1.50	QMCS	x	x
8201	32.00	33.80	1.80	Marb	x	x
8202	33.80	36.00	2.20	Fels	x	x
8203	36.00	38.00	2.00	Fels	x	x
8204	38.00	40.00	2.00	Fels	x	x
8205	40.00	42.50	2.50	Fels	x	x
8206	42.50	43.50	1.00	Arg	x	x
A0204-52	**42.50**	**43.50**	**1.00**			
8207	43.50	43.75	0.25	Fels	x	
A0204-53	**43.50**	**43.75**	**0.25**			
8208	43.75	45.70	1.95	Fels	x	x
8209	45.70	46.20	0.50	Arg	x	
A0204-54	**45.70**	**46.20**				
8210	46.20	48.00	1.80	Fels	x	x
8211	48.00	50.00	2.00	Fels	x	x
8212	50.00	52.00	2.00	Fels	x	x
8213	52.00	53.80	1.80	Fels	x	x
8214	54.25	54.60	0.35	Fels	x	x
A0204-50	**54.25**	**54.60**	**0.35**			
8215	54.60	55.70	1.10	Arg	x	
8216	55.70	56.60	0.90	Fels	x	x
8217	56.60	58.70	2.10	Arg	x	
A0204-01	**57.60**	**57.90**	**0.30**			
A0204-02	**58.25**	**58.67**	**0.42**			
8218	60.30	62.60	2.30	Fels	x	x
A0204-48	**62.60**	**63.05**	**0.45**			
A0204-03	**64.50**	**64.53**	**0.03**			
8219	65.90	67.20	1.30	Fels	x	x
8220	67.20	69.00	1.80	CS	x	x
8221	76.00	78.00	2.00	Fels	x	x
8222	78.00	80.00	2.00	Fels	x	x
8223	80.00	82.10	2.10	Fels	x	x
8224	82.90	84.10	1.20	Fels	x	x
8225	84.50	86.50	2.00	Fels	x	x
8226	91.10	92.10	1.00	Fels	x	x
A0204-55	**91.10**	**92.10**	**1.00**			
8227	92.10	94.50	2.40	Arg	x	
8228	102.40	104.30	1.90	Fels	x	x
8229	110.00	111.50	1.50	QMCS	x	x
40	samples					

Ace-02-05 Samples

ACE-02-05 Sample No.	From	To	Length	Lithology	ICP	WR
8230	7.6	10.0	2.40	Fels	x	x
8231	10.0	12.0	2.00	Fels	x	
8232	16.7	17.5	0.80	QV	x	
8233	24.7	27.0	2.30	Fels	x	
8234	27.0	29.6	2.60	Fels	x	x
5	samples					

Appendix 3
Drillhole Logs

Diamond Drill Log

DDH No. ACE-02-01

Company:	**Barker Minerals Ltd**	Date Collared	May 26, 2002
Project:	**Ace Project**	Date Completed	May 28, 2002
Logger:	Christopher J. Wild, P.Eng.	Northing	16 + 15S
Date:	May 29, 2002	Easting	6 + 65E
Remarks	Test Max-Min conductor, coincident mag, soil geochem, charg hi, resist low.	Elevation	1005 m

Azimuth	180
Dip	-45
Length	144.78 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
0.0	7.62	**Casing**	Cased through glacial till, no recovery.														
7.62	10.90	**Ovbn**	Large blocks and rounded rubble on top of bedrock, mainly quartz-musc-biotite schist, reflecting bedrock; one large block @ 8.2-8.9m contains significant garnets, not seen in bedrock here. Block was also rotated with S2 @ 10-20 to c.a.														
10.90	17.40	**QMBS**	**Quartz-Muscovite-Biotite Schist**; pale grey, slightly greenish, siliceous and schistose; consistent texture, well-foliated. Quartz-musc dominate with 5% scattered S2-parallel biotite, possibly minor chlorite giving greenish colour. S2 @ 75-85 to c.a. Weak sulphides, weakly magnetic - likely pyrrhotite (po). Very competent, weakly fractured, 100% recovery.	1									**13.0**	**14.5**	**8151**		
			13.7: 14cm white quartz vein @ 60 to c.a. (upper contact). Po-cp stwk.	7		1											
			16.75: 1-2cm layer of wispy brown massive sulphide, fine-grained and tarnished with inclusions - sulphide matrix breccia.														
17.40	19.00	**QMCS**	**Quartz-Muscovite-Chlorite Schist**; pale greenish grey, fine-grained, silceous schist, very similar to above with introduction of medium green chlorite. Biotite remains as above. S2 @ 70-80 to c.a. Weakly magnetic with finely disseminated po along S2. Continuing 100% recovery in weak to mod fractured core.	1-5									**17.4**	**19.0**	**8152**		
19.00	22.00	**QCMS**	**Quartz-Chlorite-Muscovite Schist**; dark grey, fine-grained and weakly layered, still siliceous, weakly carbonaceous on some foliation surfaces. Borderline **"argillite"**. Weakly magnetic, both pyrite and po on S2 surfaces. 100% recovery, weakly fractured. 21.8: 2-3mm wispy po band.	1	1												
22.00	26.60	**QMBS**	**Quartz-Muscovite-Biotite Schist**; same as 10.9-17.4m, weakly layered, siliceous schist, minor bitite and chlorite. S2 @ 80-85 to c.a. Po as before, weakly magnetic, perhaps f-gr pyrite along foliation planes. Weakly fractured, 100% recovery.	2	<2												
26.60	30.90	**QCMS**	**Quartz-Chlorite-Muscovite Schist**; as before, 19.0-22.0m, dark grey, fine-grained and weakly layered, still siliceous but more schistose with increased chlorite-musc, weakly carbonaceous on some foliation surfaces with increasing limy bands. Weakly magnetic, both pyrite and po on S2 surfaces. S2 @ 75-90 to c.a., locally wavy. Increased "poker chip" fracturing, loss of recovery.	1	1								**29.0**	**30.5**	**8153**		
			26.82-28.96: 70% recovery.														

Diam d Drill Log DDH No. **ACE-02-01** Date May 29, 2

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number
30.9	32.1	QMCS	**Quartz-Muscovite-Chlorite Schist;** pale green, fine-grained siliceous and musc-rich section with more limy layers. No biotite. Still minor sulphide along S2 @ 70-75 to c.a. Recovery 100%, weakly fractured.	1	<1										
32.1	39.5	Arg	**Argillite;** dark banded quartz-biotite-chlorite schist, weakly carbonaceous, less siliceous with highly contorted and brecciated sections. Locally quite calcareous with pale limy layers, and a stockwork of white calcite-quartz veinlets and stringers. Increased pyrite, less po, weakly magnetic in less deformed sections. Less competent, some loss of recovery.	<1	2-3								37.5	39.0	8154
			34.14-38.10: 80-85% recovery.												
			35.8-36.2: Less contorted qtz-chl-musc schist; S2 @ 55-65 to c.a.												
			39.0-39.5: Decreasing deformation, polished weakly graphitic slips @ 50-60 to c.a.												
39.5	46.5	QMCS	**Quartz-Muscovite-Chlorite Schist;** same as 30.9-32.1m, pale green, finer grained siliceous but with increased musc-chl and no biotite. Quartz-rich augen, 1-2%, often contain po. Increasing limy sections, generally weak. Decrease in sulphides, weakly magnetic. Locally moderately fractured.	1	<1								45.0	46.5	8155
			39.93-42.37: 80-85% recovery; core becomes less fractured @ 42.9m.												
46.5	49.0	Arg	**Argillite;** dark banded, less contorted than above (32.1-39.5m), more calcareous sections. Locally, clasts of argillite are surrounded by musc-rich schist, forming weak deformation breccia. Again, much more sulphide, mainly pyrite, only weakly magnetic. Competent, weakly fractured.	7	1								47.5	49.0	8156
			46.5: Polished, wavy graphitic slip @ 55 to c.a. marks upper contact.												
			48.4-49.0: Dark "spots", 1-2mm, hard, no form but partially post D2.												
49.0	49.7	CS	**Calc-Silicate;** pale green, with swirled not strongly schistose texture, quartz-rich, moderately calcareous patches. Well-mineralized with po, minor pyrite, moderately magnetic. Weakly fractured, 100% recovery.	7	3								49.0	49.7	8157
			49.0: Gradational upper contact from spotted musc-rich argillite.												
49.7	50.7	Marb	**Marble;** pale grey, well banded, very calcitic. Weak blebby pyrite, not magnetic. Virtually unfractured.		1								49.7	50.7	8158
			49.7: Sharp, healed fracture marks contact @ 20 to c.a. Layering @ 40 to c.a.												
			50.5: Tight fold wraps layering, both limbs @ 45 to c.a.												
			50.7: Lower contact marked by planar graphitic slip @ 30 to c.a., parallel to layering.												
50.7	52.1	Arg	**Argillite;** dark grey, well-layered and strongly contorted and folded. Sharp increase in mainly pyrite, often forming 1-2mm bands parllel to S2. More po than in previous argillite units - weak to moderate magnetism. Weakly fractured until lower contact.	5	2								50.7	52.1	8159
			51.2: S2 @ 55 to c.a., wavy.												

Diam d Drill Log DDH No. **ACE-02-01** Date May 29,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number
			51.5: S2 @ 0-70 to c.a., folded with S3 (axial plane cleavage) @ 45 to c.a. (opposite).												
			52.1: Sheared adjacent to lower contact @ 50 to c.a.												
52.1	54.0	Marb	**Marble**; as before (49.7-50.7m), tight fold @ 51.2m.	<1	<1		<<1						52.1	54.0	8160
			52.9: 5cm white quartz vein @ 70 to c.a.; minor po + sphalerite blebs.	2			3								
			53.3: Polished graphitic slip @ 60 to c.a.												
			54.0: Increasing musc-chl, sharp drop in calcite.												
54.0	55.2	QMCS	**Quartz-Muscovite-Chlorite Schist**; pale greenish grey, fine-grained, very siliceous, hard, weakly fractured. Quite contorted weak foliation, quartz veining common. Up to 5% disseminated po, moderately magnetic, less pyrite.	5	1								54.0	55.2	8161
			54.9: S2 @ 55 to c.a. in dark argillite section.												
55.2	58.8	Arg	**Argillite**; same as above, dark, moderately graphitic, mainly along S2 slips, very siliceous, not calcareous. Increase in pyrite, decreased po, weak to mod magnetic. Competent except for shear zone near bottom.										55.2	57.0	8162
													57.0	58.8	8163
			55.7: Bands of wispy py-po, along S2 @ 75 to c.a.												
			56.0: Bands of wispy py-po, along S2 @ 45 to c.a.												
			57.4-57.7: **Fault**; quartz-graphite, 5-10cm of graphitic gouge with white quartz vein.												
			57.7-58.5: **Fault (con't)**; strongly sheared, becoming brecciated, numerous graphitic slips and minor gouge.												
			58.7: Pyrite flattened along S2 @ 80-90 to c.a.												
58.8	65.3	Fels	**Felsite**; pale grey, conspicuously mottled unit composed of very fine-grained plagioclase with minor but variable biotite (reddish flecks) and muscovite. Locally calcareous patches, notable around disseminated to blebby po - moderately magnetic. Unit is dense, hard, very competent with 100% recovery.	5-10	1								58.8	61.0	8164
													61.0	63.0	8165
													63.0	65.3	8166
			58.8: Fuzzy but distinct upper contact @ 85 to c.a.												
			61.2: Blebby sphalerite in quartz vein.												
			63.6-63.9: **Quartz-Muscovite-Chlorite Schist**; dark siliceous schist. S2 @ 85 to c.a.												
			64.4-64.5: 4cm layer of grey **Marble**, @ 85 to c.a.												
			65.3: Becomes spotted, possibly dark quartz eyes.												
65.3	67.5	QCMS	**Quartz-Chlorite-Muscovite Schist**; darker, moderately foliated, siliceous schist, possibly fine-grained biotite or graphite. At least 5% po + py, much of it as wispy layers. Weakly fractured.	5	2								65.3	67.5	8167
			65.4-65.5: 10% wispy po bands @ 80 to c.a.												
			66.2: 5cm band of 20% po, minor py.												
			67.5: Sharp lower contact, S2 @ 80 to c.a.												

Diam[illegible]d Drill Log DDH No. ACE-02-01 Date May 29, [illegible]

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
67.5	71.2	QMBS	**Quartz-Muscovite-Biotite Schist**; pale grey, slightly greenish, siliceous and schistose; consistent texture, well-foliated. Quartz-musc dominate with 5% scattered S2-parallel biotite, possibly minor chlorite giving greenish colour. S2 @ 75-85 to c.a. Mod f-gr sulphides, weakly to mod magnetic - mainly pyrrhotite (po). Very competent, weakly fractured, 100% recovery.	2-3	1												
			67.5: Sharp, unsheared contact @ 80 to c.a., same as S2.														
			71.2: Lower contact gradational over 15cm.														
71.2	79.5	QMCGS	**Quartz-Muscovite-Chlorite-Garnet Schist**; green, fine-grained siliceous schist with strong musc-chl, minor flecks of biotite, and 2% pink porphyroblastic garnets. A few musc-rich sections are weakly sheared. Weaker fine-grained po.	2-3	<1												
			72.7: Minor 10cm shear zone, S2 @ 75 to c.a.														
			73.7-74.0: Minor shear zone.														
			75.5-76.1: Minor shear zone @ 20-30 to c.a., S2 @ 60 to c.a.														
			76.3-76.4: Strong pink garnets to 5mm diameter.										76.0	77.5	8168		
			78.5: 5-10cm shear zone, S2 @ 55 to c.a.														
			79.5: Weak shear along S2 @ 80 to c.a., marks contact, last of garnets.														
79.5	86.9	QMCS	**Quartz-Muscovite-Chlorite Schist**; strongly micaceous unit, broken up; similar to above unit but lacking garnets. Somewhat darker and more finely layered than above. Sheared zones show good crenulations on S2. Weak sulphides, weakly magnetic, py + po, disseminated and stringer veinlets.	2	1												
			80.2: S2 @ 80 to c.a.														
			81.7: Weak shear zone, S2 @ 50 to c.a.														
			81.69-83.52: 65% recovery, lost ~83.5m, in poker chip schist.														
			83.5-86.9: Well-fractured but not extensively sheared, some poker chips.														
			85.5: S2 @ 75 to c.a.														
86.9	93.5	QMCGS	**Quartz-Muscovite-Chlorite-Garnet Schist**; green (slightly paler than above unit), fine-grained siliceous schist with strong musc-chl, minor flecks of biotite, and resumption of 2% pink porphyroblastic garnets. A few musc-rich sections are weakly sheared. Weaker fine-grained po.	2-3	<1								88.0	89.5	8169		
			89.0-89.2: Strongly garnetiferous, irregular S2 parallel quartz veining.														
			89.2-89.3: Musc-rich shear zone, S2 @ 80 to c.a.														
			90.3-90.9: Musc-rich shear zone, S2 @ 90 to c.a.														
			91.4: 15cm barren white quartz vein, upper contact @ 80 to c.a.														
			93.0: S2 @ 75 to c.a.														
93.5	102.0	QCMS	**Quartz-Chlorite-Muscovite Schist**; darker, moderately foliated and finely layered, siliceous schist, possibly fine-grained biotite or graphite, and no garnets; not calcaeous. At least 5% po + py, much of it as wispy layers, moderately magnetic. Weakly fractured.	5	2												

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
			95.71-97.54: 45% recovery in soft micaceous and locally gougy shear zone.														
			99.0: S2 @ 75 to c.a., consistent.														
			100.7: S2 @ 75 to c.a., unit is very consistent in texture and orientation.														
102.0	103.6	**QMCGS**	Same 86.9-93.5m, paler green, less strongly foliated and layered, presence of scattered pink garnets, less po + py, less magnetic. S2 @ 75-80 to c.a.	**2**	**<1**												
103.6	112.7	**QCMS** (Arg)	**Quartz-Chlorite-Muscovite Schist**; darker, moderately foliated and finely layered, siliceous schist, possibly fine-grained biotite or graphite, and only occasional ragged garnet porphyroblasts; not calcaeous. At least 5% po + py, much of it as wispy layers, moderately magnetic. Weakly fractured to 107.4, then moderately fractured.	**3**	**1**								**103.6**	**105.1**	**8170**		
			103.7: Ragged garnet porphyroblasts, one 12mm in diameter, classic augen.														
			107.0: S2 @ 75 to c.a.														
			107.4-112.7: Moderately fractured, much of it to "poker chips"; 75-80% recovery.														
			110.5: Minor gouge along foliation.														
			111.3: S2 @ 80 to c.a., very planar.														
112.7	132.8	**QMCGS**	**Quartz-Muscovite-Chlorite-Garnet Schist**; green (slightly paler than above unit), fine-grained siliceous schist with strong musc-chl, minor flecks of biotite, and resumption of 2% pink porphyroblastic garnets. A few musc-rich sections are weakly sheared. Weaker fine-grained po.	**2-3**	**<1**								**130.0**	**131.5**	**8171**		
			113.9: S2 @ 65 to c.a., wavy and crenulated.														
			112.47-114.30: 60-65% recovery, losses at both ends.														
			114.30-117.35: 70% recovery, losses around 114.3m.														
			114.3-115.7: 35% recovery; much white quartz vein rubble in strongly garnetiferous musc-rich phyllite; likely a **fault zone**. 12cm of crowded garnet, quartz vein material, musc-rich phyllite with strong kink folds (M-type) with S3 @ 45 to c.a., perpendicular to S2.														
			117.35: S2 @ 70 to c.a., planar.														
			120.4: S2 @ 75 to c.a.														
			102.8-122.0: Moderately fractured, some poker chips.														
			122.4-123.2: Scattered biotite; qtz-musc-biotite schist.														
			122.7: 3-4cm white quartz vein @ 70 to c.a.; 10% po, 1-2% py in fractures.														
			123.7-124.4: Weak qtz-musc-biotite schist, less chloritic.														
			124.4-124.8: Green chloritized porphyroblasts, after garnet?														
			124.1: 1-2cm white quartz vein, 5% po, 5% py.														
			125.9-126.7: Poker chip fracturing.														
			127.1-138.0: Mod to strongly fractured; 127.9: 0.5cm white qtz vein @ 70 to c.a.														
			129.6: Irregular 1-3cm thick barren white quartz vein.														
			131.0: 1cm white quartz vein, 20% po, 4% py, 1% cp, along fractures in vein @ 80 to c.a.														

Diam d Drill Log DDH No. **ACE-02-01** Date May 29,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
			132.0: S2 @ 80 to c.a, weakly crenulated, sericitic.														
132.8	135.0	Arg	**Argillite / Quartz-Graphite Schist**; dark, well-layered moderately graphitic schist, focus of moderate faulting and associated white quartz veining. Not extensively fractured and only minor gouge, polished graphite along several S2 partings and breccia at lower contact.	<l	1												
			132.8: Irregular crosscutting white quartz vein with patches of tan carbonate, coarse bleb of pyrite, no po noted here.														
			132.9: Polished graphite slip @ 70 to c.a.; patch of brown mineral with up to 20% po over 3-4cm - moderately magnetic.														
			133.0-133.6: Highly contorted, folded quartz-graphite schist.														
			133.20-134.11: 70% recovery.														
			134.0-134.1: **Fault**; pale greenish clay gouge, some loss of recovery.														
			134.1-134.2: White quartz vein bound by polished graphitic slips @ 75 to c.a. (erratic).														
			134.6-134.9: Quartz-graphite shear zone, numerous polished graphite slips.														
			134.9-135.0: Fault breccia; subrounded to subangular clasts in clay/graphite matrix; lower slip @ 50 to c.a.														
135.0	144.8	QMCS	**Quartz-Muscovite-(Chlorite) Schist**; pale green, weakly layered and mainly waxy looking like sericite, soft but weakly fractured. Blebby po associated with quartz veining, but overall weakly mineralized with po + py.	1	1								**139.5**	**141.0**	**8172**		
			136.0: Thin (1-2mm) clay gouge slip.														
			137.6: Quartz vein, 2-3cm thick, @ 70 to c.a.; 2-5% magnetic po wisps.														
			138.8: Irregular quartz veining, with 50% patchy to stringer dark brown carbonate (mod rx with cold HCl), and 5% associated po.														
			139.8-140.7: Increased quartz flooding with many distinct white quartz veins intruding pale green schist. No sulphides discerned.														
			141.8: Flattened py + po along S2 @ 80 to c.a.														
			143.6: 3-4cm white quartz vein with blebby po along selvages and minor py, min po along tension cracks perpendicular to the vein; @ 65 to c.a														
			143.6-143.8: Qtz-muscovite-biotite schist.														
			143.9-144.2: 5% blebby to stringer bronze, magnetic po.														
			144.3-144.4: 7cm white quartz vein with minor py along selvage.														
			END OF HOLE = 144.78 metres														

Diamond Drill Log

DDH No. **ACE-02-02**

Company:	**Barker Minerals Ltd**	Date Collared	May 29, 2002
Project:	**Frank Creek**	Date Completed	May 31, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	18 +80S
Date:	May 31, 2002	Easting	8 + 64E
Remarks	Test mag high at L19S, 7+25E to 8+00E	Elevation	1025 m

Azimuth	180
Dip	-45
Length	164.90 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
0.00	13.72	**Casing**	Cased through road fill and glacial till, no recovery.														
13.72	39.40	**QMCGS**	**Quartz-Muscovite-Chlorite-Garnet Schist**; pale greenish grey, siliceous and schistose; consistent texture, well-foliated. Quartz-muscovite-chlorite dominate, quite soft, with 2% scattered porphyroblastic pink garnets typically 1mm diameter. S2 @ 75-85 to c.a. Weak to moderate sulphides, weakly magnetic - pyrrhotite (po), pyrite (py), and chalcopyrite (cp); concentrated along S2 planes. Very competent, weakly fractured, 100% recovery.	**<1**	**1**	**<1**							**26.0**	**27.5**	**8173**		
			13.7-14.9: Well contorted siliceous bands, garnets associated with micaceous layers.														
			15.2-15.8: 5+% spotted brown biotite in qtz-musc-bi schist; no garnets.														
			15.5: 6cm quartz vein, well-mineralized with blebby and stockwork pyrrhotite, lesser pyrite and/or marcasite, and traces of fine-grained cp.														
			18.3: S2 @ 75 to c.a.														
			21.2: Weakly gougy slip along S2 @ 70 to c.a.														
			21.3: Medium-grained pyrite and irregular fracture.														
			21.8: Minor weakly magnetic po + cp on S2 @ 80 to c.a.														
			22.8: Minor weakly magnetic po + cp on S2 @ 75 to c.a.														
			24.1: Minor py + cp on S2, adjacent to wispy 2-3cm barren white quartz veining.														
			25.1-26.0: Weak spotted biotite in QMCGS.														
			30.4: Weak calcite, minor py-po, trace cp on S2 @ 85 to c.a.														
			32.0: Minor po-py-cp on S2 @ 80 to c.a.														
			35.0: Minor po-py-cp on S2 @ 80 to c.a.														
			36.0: 3-4cm wispy white quartz vein; po>py>cp along selvages, S2.														
			39.4: Sharp contact along S2 @ 80 to c.a.														
39.40	41.70	**Marb**	**Marble**; darker grey, less green, fine-grained, finely layered, dirty marl, quite chloritic possibly > musc. Calcitic with strong reaction to HCl. Continues competent and weakly fractured. Less sulphides po>py.	**<1**	**<1**												
41.70	46.80	**QMCGS**	**Quartz-Muscovite-Chlorite-Garnet Schist**; medium to greenish grey, fine-grained, siliceous to micaceous schist, occasional garnets and weakly carbonaceous. S2 @ 70-80 to c.a. Weakly magnetic with finely disseminated po along S2. Continuing 100% recovery in weak to moderately fractured core.														

Diam d Drill Log DDH No. **ACE-02-02** Date May 31, 2

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
			42.9: Irregular quartz vein with patchy brown carbonate (tourm?) and blotchy po, 1-2cm.														
			44.0: Kink folds, S2 is axial planar cleavage, layering folded, begin fold zone.														
			44.7: 16cm white quartz vein with cream patches of carbonate and minor <5%, blebs of po. Lower contact @ 35 to c.a.														
			44.8-46.8: Increase in po, less py, slightly more magnetic; finely disseminated.														
			45.8: S2 @ 40 to c.a.; continuing quite contorted.														
			46.5-46.8: Well-fractured, po>py in white quartz vein along contact @ 45 to c.a.														
46.8	47.7	QMBS	**Quartz-Muscovite-Biotite Schist;** pale grey, brownish speckled, musc-rich schist; bands of biotite, weak chlorite often with pink garnets. Decreasing fracturing - very competent. Weakly disseminated po-py.														
47.7	62.4	QMCS	**Quartz-Muscovite-Chlorite Schist;** darker grey, well-layered, more strongly chloritic than in musc-rich, garnet schist; no obvious garnets. Significant po-py, minor cp along weakly polished S2 planes. Weakly fractured, quite competent.	2	<1	tr											
			47.9-48.1: Irregular quartz vein, upper contact @ 15 to c.a., lower @ 60 to c.a Quartz is mottled with coarse patchy brown mineral, reacts weakly with HCl but may be fine-grained tourmaline. 10-20% po>py, minor cp.										47.9	48.1	8174		
			48.4: Layering is crenulated with a strong S2 cleavage @ 50 to c.a.														
			48.9: More open fold; vergence is uphole, to north.														
			50.0-50.8: Darker with fine felted texture, possibly tourmaline.										50.0	51.5	8175		
			55.8: White, 3-5cm quartz vein, 5% po along selvage, minor greenish muscovite-sericite dispersed through vein, diffuse boundaries, with S2 @ 85 to c.a.														
			56.4: White quartz vein, 14cm thick, @ 70-75 to c.a., minor py along fractures.														
			57.3: Small tight folds, verging uphole to north (to antiformal closure).														
			58.4: Po, minor cp along S2 @ 90 to c.a.														
			59.7-60.1: Strong "poker chip" fracturing.														
			60.8: Py-po-minor cp on S2 @ 90 to c.a.														
			62.4: Sharp lower contact, parallel to S2 @ 85 to c.a.														
62.4	88.8	QMCGS	**Quartz-Muscovite-Chlorite-Garnet Schist;** pale greenish grey, siliceous and schistose; consistent texture, well-foliated, commonly contorted. Quartz-muscovite-chlorite dominate, quite soft, with 2% scattered porphyroblastic pink garnets typically 1mm diameter. S2 @ 75-85 to c.a. Weak to moderate sulphides, weakly magnetic - pyrrhotite (po), pyrite (py), and chalcopyrite (cp); concentrated along S2. Very competent, weakly fractured, 100% recovery.	2	<1												
			66.3: Coarse very ragged, pink garnets, grass green chlorite in pressure shadows.														
			66.5: Open fold, vergence uncertain, possibly D3.														
			68.6: S2 @ 75-90 to c.a., py>po on some S2 planes.														

Diam d Drill Log   DDH No.   ACE-02-02   Date   May 31,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
			71.8-73.7: Strong "poker chip" fracturing; strong muscovite, py>po.														
			75.4: S2 @ 80 to c.a.														
			79.8: 3-4cm of pale grey quartz eye section with wispy grass green chlorite. Flattened pyrite along several S2 planes @ 75 to c.a.										78.5	80.0	8176		
			81.0: S2 @ 70 to c.a.														
			82.5-83.3: **Marble**; dirty, dark grey, highly schistose and well-layered. Interlayered with minor pale greenish QMCGS. Moderate to strongly calcareous layers. Layering (S2) @ 80 to c.a., uniform.														
			84.6-85.7: **Marble**; same as 82.5-83.3m.														
			87.1-87.2: **Marble**; same as 82.5-83.3m.														
			88.1-88.45: **Marble**; same as 82.5-83.3m.														
88.8	92.4	Marb	**Marble**; same as 82.5-83.3m, fairly uniform, mod to strongly calcareous, with obvious calcitic layers and darker chl-musc layers (marl). Fine-grained po throughout, weak to moderate magnetism. Weakly to moderately fractured. 89.9-90.0: S2 @ 75-80, S3 @ 50-55 to c.a.	2-3	<1												
92.4	105.9	QMCGS	**Quartz-Muscovite-Chlorite-Garnet Schist**; same as above (62.4-82.5m). Considerably more fractured, esp around minor gougy faults. Mineralization consists of po finely disseminated throughout and pyrite as flattened grains along S2 fracture planes.	1-2	1-2												
			94.7-97.3: Sharp increase in fracturing, some "poker chips".														
			96.0-96.1: **Fault**; moderate intensity, little displacement; lower slip @ 20 to c.a.														
			99.7-100.0: "Poker chip" fracturing; S2 @ 85-90 to c.a.														
			101.4: S2 contorted, S2 @ 25 to c.a.														
			101.7-101.8: Intense "poker chip" fracturing.														
			102.0: Quartz vein breccia, approx 5cm thick, @ 25 to c.a.; 3% py, 3% po, black stockwork - possibly tourmaline.														
			102.1-105.9: Strong contortion of layering and S2, incr irregular quartz veining, minor "poker chip" fracturing. Decr garnets and increase in blebby pyrite.	<1	2-3												
105.9	108.5	Flt	**Fault**; assortment of quartz-graphite and clay gouge zones with sharp increase in carbonate and graphite.														
			105.9-108.0: Mainly quartz-graphite schist ("argillite"), polished graphitic slips.														
			107.3: Dark grey clay-graphite-quartz gouge, 5cm thick.														
			108.0: Lower graphitic slip @ 30 to c.a.														
			108.2-108.5: Mainly pale greenish gouge; wavy lower "contact", polished clay-sericite @ ~90 to c.a.														

Diam d Drill Log DDH No. **ACE-02-02** Date May 31,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number
108.5	118.8	QMCS	**Quartz-Muscovite-Chlorite Schist**; pale green, fine-grained and more massive-looking, alternating harder and more siliceous with softer, more micaceous schist. Strongly fractured with white sericite coating the core to 122.7m. Mineralization includes pyrite on S2 fractures and po disseminated throughout, both somewhat weaker than before.	1	1										
			110.0-111.1: Zone of silicification surrounding a brittle white quartz vein, 10-15cm thick @ 110.5m. Vein is locally vuggy, hosts 5% pyrite. Silicified section hosts several wispy white quartz veins in siliceous muscovite schist. S2 and veins are 45-60 to c.a.										110.0	111.5	8177
			112.5: 4-5cm section of pale clay gouge @ 80 to c.a.												
			112.7: S2 folded around fairly open fold - no vergence.												
			113.8-114.8: Strongly silicified, as before (110.0-111.1m).												
			116.2: Minor gouge on S2 fractures.												
			116.3-117.5: Moderately to strongly silicified zone, as above.												
			116.8-188.8: Strong fracturing.												
118.8	122.5	Fault	118.57-119.18: 65% recovery.												
			118.8: Upper slip @ 35-45 to c.a.												
			118.8-119.3: Mainly pale green gouge and rubbly fault breccia.												
			119.6-120.4: Strongly sheared with soft sericitic fractures in vuggy schist @ 60 to c.a.												
			120.40-121.31: 20% recovery, mainly pebbly rubble.												
			121.31-122.53: 65% recovery,												
			121.5 (approx)-122.5: Mainly pale green gougy matrix and breccia blocks.												
122.5	164.9	QMCS	**Quartz-Muscovite-Chlorite Schist**; as above, 108.5-118.8m; somewhat less fractured and less quartz veining. Minor calcite on late fractures.	1	1										
			122.8: White quartz vein with very diffuse boundaries, 5-6cm thick.												
			123.3: S2 @ 65 to c.a.												
			123.4-123.8: Similar diffuse quartz veins, aligned with S2, minor magnetic po, py.	2	<1										
			128.0: Muscovite-rich, minor shear; S2 @ 65-70 to c.a.												
			128.4: White quartz vein, 1-3cm thick with 1 fracture-controlled bleb of po.												
			129.0-130.8: Strong silicification, diffuse white quartz veining, only minor sulphides.												
			131.8-132.4: White quartz vein, minor intervening musc schist. Upper contact @ 55 to c.a., along S2. Hosts large patch of fine-grained po mixed with pyrite; reddish-bronze mineral may be sphalerite. Fractured at lower contact.	7	5	tr	1						131.8	132.5	8178
			134.1: 1cm white quartz vein with 10% po + py, minor cp. Vein @ 55 to c.a.												
			135.8-136.2: Strongly silicified zone, wispy musc schist @ 30 to c.a.												
			136.9: Quartz vein with strong planar graphite slip @ 48 to c.a.												
			138.5: S2 @ 70 to c.a.												
			138.8: Lensoidal to blebby po 1-2mm thick along S2 @ 70 to c.a.												

Diam d Drill Log DDH No. **ACE-02-02** Date May 31,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Au ppb
			141.4: 1-2cm white quartz vein @ 50 to c.a.; brown patchy mineral - tourmaline?, minor py, lesser po.														
			144.3: Blebby py + cp, roughly along S2 @ 80 to c.a.; weak to mod calcareous.														
			145.1-146.4: Increased fracturing, reflects more micaceous character.														
			147.7: 3cm white quartz vein @ 55 to c.a., po along selvages and fractures.														
			147.8: S2 @ 85 to c.a.														
			155.2: Begin to see some darker layers in pale green schist. Also, 2-3cm white quartz vein with patchy dark brown mineral with po+py, esp adjacent to lower contact.										151.0	152.5	8179		
			155.8-160.2: Darker, more schistose, moderately to well-fractured, local "poker chips", minor shearing.														
			158.1-158.2: White quartz veining with polished graphitic slips @ 60 to c.a. Unsheared lower contact @ 60 to c.a., tight fold against vein.														
			159.6: White quartz vein, 4-5cm thick @ 50-70 to c.a., brown patchy mineral + po > py.														
			160.9-163.6: Numerous white quartz veins, fuzzy silicification.														
			161.24-163.07: 67% recovery, no obvious cause of core loss.														
			163.8: Minor shear zone, musc-rich gouge over 5cm.														
			163.9-164.9: Back to pale greenish, more massive-looking schist.														
			164.8: Uniform kink folds with axial planes @ 55-60 to c.a., verging to antiform uphole.														
			END OF HOLE = 164.90 metres														

Diamond Drill Log

DDH No. **ACE-02-03**

Company:	**Barker Minerals Ltd**	Date Collared	May 31, 2002
Project:	**Frank Creek**	Date Completed	June 2, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	4 + 70N
Date:	June 2, 2002	Easting	5 + 70W
Remarks	Drill fence with DDH-98-3, test mineralization in trench TR-97-1B	Elevation	1028 m

Azimuth	180
Dip	-45
Length	150.88 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
0.00	8.23	**Casing**	Cased through fill from trench and till, no recovery.														
8.23	55.50	**Fels**	**Felsite**; pale grey, conspicuously mottled unit composed of very fine-grained plagioclase with minor but variable biotite (reddish flecks) and muscovite. Locally calcareous patches, notable around disseminated to blebby po - moderately magnetic. Unit is dense, hard, weakly fractured, and very competent with 100% recovery.	**5-7**	**3**												
			8.2-8.5: A few cobbles at bedrock interface.														
			8.5-9.7: Semi-Massive Sulphides; approx 20-30% pyrrhotite and pyrite; po forms a distinctive net-texture - very fine grained while pyrite is later, coarser grained and forms within 1-2cm layers. Moderately to strongly magnetic.	**20**	**10**								**8.5**	**9.7**	**8180**		
			9.7-11.8: Layered and Disseminated Sulphides; approx 15-20% po +pyrite; pyrite more flashy, coarse-grained and in layered, as above, but much less fine net-texture po. Sulphide layers and S2 @ 80 to c.a for most of interval.	**5**	**10**								**9.7**	**11.8**	**8181**		
			9.7-10.7: Well-fractured, weakly gougy or rubbly.														
			11.8-13.9: Well-layered, micaceous section, with several 2-3cm layers of 25% coarse pyrite, minor po, possibly traces of reddish sphalerite.										**11.8**	**13.9**	**8182**		
													13.9	**16.0**	**8183**		
			13.8: White quartz vein, 2-3cm thick, @ 25-30 to c.a. 10% coarse pyrite with several coarse grains of reddish sphalerite.										**16.0**	**18.0**	**8184**		
													18.0	**19.6**	**8185**		
			13.9-14.6: Pale grey to white, very fine-grained and siliceous-looking; py > po in weak stockwork, disseminations and blebs, weakly magnetic.														
			14.6-15.9: Well-layered and darker grey, esp 15.2-15.8m, becomes more cherty in character. Decrease in sulphides.														
			15.8: S2 @ 55 to c.a.														
			17.2-17.5: Healed breccia, likely tectonic.														
			17.5: Musc-rich gougy section around layer of py-po and brown tourmaline (?).														
			17.5-19.6: Layered pale grey, few micas, <5% sulphides.														
			19.6-35.5: Weakly spotted, pale grey to white, very uniform, weak micas so no real foliation developed. 5-10% po > py, moderately magnetic. Very weakly fractured.				**tr**						**19.6**	**21.0**	**8186**		
			21.6-21.7: More biotite-rich section.														
			22.6-22.9: White quartz vein, 2cm thick, blebby po, flattened py along fractures, @ 20 to c.a.														
			22.9: S2 @ 70 to c.a., weak.														

Diam d Drill Log DDH No. **ACE-02-03** Date June 2,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number
			26.9-27.4: Stockwork brecciation, autoclastic.										26.9	28.3	8187
			28.3-30.0: Looks glassy or like polished marble, minor diss sphalerite with py.										28.3	30.0	8188
			31.3-31.4: 25cm milky white quartz vein, sharp contacts, lower @ 30 to c.a.												
			32.1-33.6: Blocky breccia, healed but vuggy between blocks; locally looks like a lapilli tuff with fine py-po stockwork forming the matrix.												
			35.5-38.6: Brownish speckled biotite, possibly an alteration - 10-20% biotite with coarse pyrite grains superimposed on foliation.												
			37.7: S2 @ 70 to c.a.												
			39.5-39.8: Biotite-rich section, as before.												
			39.8-43.3: Layered pale greenish and weaker biotite sections, less felsic.										40.0	41.5	8189
			42.9-43.3: Milky white quartz vein, minor po, py, traces of cp; vein @ 70 to c.a.												
			43.3-50.3: Weak but consistent biotite-felsite, 10% biotite; fine-grained disseminated po > py, associated with biotite, parallel to S2.												
			<u>50.3-51.3</u>: **<u>Quartz-Muscovite-Chlorite Schist (QMCS)</u>**; pale green, fine-grained, massive-looking.										50.3	51.3	8190
			51.3-55.5: Weak biotite-felsite, as before.												
55.5	57.7	**Arg**	**<u>Siliceous Argillite</u>**; dark grey banded, strongly siliceous pale and dark grey layers, strongly contorted. Sharp upper contact @ 75 to c.a., parallel to S2. Gradational lower contact, decreasing dark layers into felsite. Slight decrease in sulphide content, weak disseminated po + py. More fractured than felsite but hard and competent.	1	1								55.5	57.7	8191
57.7	59.6	**Fels**	**<u>Felsite</u>**; pale grey, conspicuously mottled unit composed of very fine-grained plagioclase with variable biotite (reddish flecks) and muscovite. Locally calcareous patches, notable around disseminated to blebby po - moderately magnetic. Unit is dense, hard, weakly fractured, and very competent with 100% recovery.	3-4	1								57.7	59.6	8192
59.6	63.6	**QMCS**	**<u>Quartz-Muscovite-Chlorite Schist</u>**; pale green, fine-grained, homogeneous. Relatively soft yet weakly fractured. Fine-grained disseminated po, minor py.	2	<1										
63.6	70.8	**Fels?**	**<u>Quartz-Biotite-(Muscovite-Chlorite) Schist or Felsite</u>**; very distinctive biotite-rich fine-grained siliceous or felsic (plagioclase)-rich unit in which biotite forms thin bands often folded and swirled. Locally finely layered, resembles banded chert in places. Also resembles more biotite-rich section of felsite unit. Very weakly magnetic, weak po, increased coarser late pyrite.	1	2								64.0 69.3	65.5 70.8	8193 8194
70.8	72.6	**QMCS**	**<u>Quartz-Muscovite-Chlorite Schist</u>**; pale green, fine-grained, homogeneous. Relatively soft yet weakly fractured. Fine-grained disseminated po, minor py.	2	<1										

Diam[illegible]d Drill Log DDH No. **ACE-02-03** Date June 2, [illegible]

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
72.6	78.9	Fels	**Felsite**; pale banded brown and grey, weakly mottled unit composed of very fine-grained plagioclase with variable up to 5% biotite (reddish flecks) and muscovite-rich sections. Locally calcareous; disseminated to blebby po - moderately magnetic and py, both with weak S2, py and po-rich sections, less overlap. Unit is dense, hard, weakly fractured, and very competent with 100% recovery.	3-4	3-4								72.6	74.6	8195		
			78.9: Sharp lower contact @ 60-65 to c.a., parallel to S2.														
78.9	81.7	QMCS	**Quartz-Muscovite-Chlorite Schist**; pale green, weakly schistose, quite siliceous but relatively soft, possibly with chlorite. Same as 70.8-72.6m.	2	<1												
			80.8: Minor white quartz vein, 2-3cm thick with a 5mm bleb of po.														
			81.2: Felsite: 14cm zone of strongly mineralized (blebby po+py - 10%), quite micaceous, S2 @ 70 to c.a.														
			80.6-81.7: Moderately fractured, increased to contact.														
81.7	95.1	Arg	**Siliceous Argillite**; mixed unit of QMCS, weakly to strongly carbonaceous with local graphitic slips. Significantly darker and well-layered compared to surrounding muscovite-rich schists, weakly to moderately calcareous. Continuing diss but weaker py+po, weakly magnetic. Moderately fractured.	1-2	1-2												
			83.5-83.8: QMCS, siliceous.														
			84.1-84.7: Marble; mod to strongly calcareous, musc-rich bands. S2 @ 65-75 to c.a.														
			84.7-85.9: Quartz-Graphite Schist / Fault; Moderately calcareous, dark grey layered with numerous curved and polished graphite slips, minor white quartz veining. Well-fractured, only 50% recovery.														
			85.9-89.8: Well-layered, dark and pale grey to white, 1-3mm "stripes", with occasional 1-2mm pyrite bands; more competent.														
			86.0: Needle-like mineral, up to 5mm long, usually with S2 but clearly post D2.														
			89.8-90.2: Quartz-Muscovite-Chlorite Schist; pale green, siliceous unit, verging on felsite. Significant wispy po, blebby later py. Sharp upper contact @ 70 to c.a., flatter, more contorted lower contact.														
			90.2-91.5: Contorted dark banded schist.														
			91.5-91.7: More needle-like crystals, clearly crossing S2. (Leucoxene?)														
			91.7-93.0: Well-fractured, rubbly including white quartz pebbles, sharp loss of recovery - likely a fault centred around 92.0-92.5m.														
			89.61-92.66: 70-75% core recovery.														
			92.66-95.71: 60% core recovery.														
			94.3-95.1: Gradational lower contact, becoming pale green, musc-rich.														
			95.0: S2 @ 70 to c.a.														

Diam d Drill Log DDH No. ACE-02-03 Date June 2,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
95.1	106.2	QMCS	**Quartz-Muscovite-Chlorite Schist**; pale green, fine-grained and becoming more massive-looking, alternating harder and more siliceous with softer, more micaceous schist. Mineralization includes pyrite on S2 fractures and po disseminated throughout, both somewhat weaker than before.	1	1								95.1	97.0	8196		
			98.8: S2 @ 77 to c.a., planar and consistent.														
			102.5-104.2: Moderately fractured, no evidence of brittle shearing.														
			103.2: 2-3cm layer of darker grey banded **limestone**, first appearance.														
			103.5: Tight fold outlined in core - compositional layer, well-developed axial planar cleaveage (S2) @ 60 to c.a. - no vergence.														
			103.7-104.0: Dirty grey banded limestone or marl; moderate reaction with dilute HCl. Layering, S2 @ 75 to c.a., tight fold at upper contact (F2).														
			104.4: Limestone band, very irregular fold.														
			105.4, 105.6: Two limestone layers, as above.														
106.2	107.7	Marb	**Marble**; medium grey, well-banded, somewhat contorted, moderately calcareous, dirty limestone or marl. Quite micaceous - chlorite and muscovite. Weakly mineralized, fine-grained pyrite, not magnetic. Weakly fractured, very competent.	<<1	<1												
107.7	113.9	QMCS	**Quartz-Muscovite-Chlorite Schist**; pale green, fine-grained and becoming more massive-looking, alternating harder and more siliceous with softer, more micaceous schist. Mineralization includes pyrite on S2 fractures and po disseminated throughout, both somewhat weaker than before.	2	<1	<1											
			107.7-: 5% pale to white tabular to needle-like crystals (leucoxene?)														
			110.5: Bronze po with minor cp and py, on fracture (S2).														
			111.3: S2 @ 80 to c.a.														
			113.9: Lower contact, S2 @ 70-75 to c.a.														
113.9	131.3	Arg	**Argillite**; darker green-grey, fine-grained and well-layered but strongly contorted and folded; continuing quite siliceous. Sharp increase in secondary coarser-grained pyrite, minor fine-grained lenticular po. Needle-like white crystals still common.	1	3												
			116.9-117.2: Tightly folded, micaceous, well-fractured, weak shear zone.														
			117.3: S2 @ 70 to c.a.														
			118.2-118.4: Well-fractured, micaceous, strongly crenulated S2 surfaces.										118.0	119.2	8197		
			120.4: S2 @ 65 to c.a.; folding evident but not measureable.														
			125.4-125.9: Fault; 5cm dark clay gouge, weakly graphitic, followed by 45cm of "poker chips", numerous polished micaceous slips.														
			125.9-126.5: Quartz-Muscovite-Chlorite Schist; pale green, fine-grained, gradational with darker schist.D22														
			126.5-128.0: Darker, quartz-chlorite-muscovite schist, argillite.														
			128.0-128.5: QMCS, as above.														

Diamond Drill Log   DDH No. ACE-02-03   Date June 2,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
			128.3: 10cm white quartz vein, upper contact @ 35-40 to c.a.; minor pyrite.														
			128.9-129.5: Series of weak S2-parallel shear zones, micaceous.														
			131.1: 10-12cm white quartz vein, parallel to S2 @ 70 to c.a., micaceous partings.														
			131.3: irregular, weakly gougy fault @ 30 to c.a., schist same across fault, contact parallel with S2 @ ~70-75 to c.a.														
131.3	150.9	QMCS	<u>**Quartz-Muscovite-Chlorite Schist**</u>; pale green, fine-grained weakly foliated and more massive-looking, quite siliceous and monotonous unit. Occasional white quartz veins, 1-2cm thick, largely lacking sulphides. Mineralization includes minor coarse pyrite occasional disseminated po, both somewhat weaker than before. Very weakly magnetic. Very competent and weakly fractured.	<1	<1								132.0	133.5	8198		
			136.2: S2 @ 80 to c.a.														
			140.8: Weakly gougy fracture, @ 30 to c.a.														
			142.6: S2 @ 80 to c.a.														
			145.8: S2 @ 75 to c.a.														
			147.2-147.6: Weakly gougy (micaceous) S2-parallel shear zone.														
			148.3-149.3: Series of weakly gougy, S2-parallel shear zones.														
			149.7: Somewhat darker and more chloritic to end, folds evident but not measured. Increase in blebby lenticular pyrite.														
			150.8: White quartz vein, 6cm thick @ 60-70 to c.a. (parallel to S2), patchy tan carbonate, muscovite and dark brown, felted tourmaline along lower selvage. No sulphides detected.														
			END OF HOLE = 150.88 metres														

Diamond Drill Log

DDH No. **ACE-02-04**

Company:	**Barker Minerals Ltd**	Date Collared	June 2, 2002
Project:	**Frank Creek**	Date Completed	June 3, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	4 + 20N
Date:	June 4, 2002	Easting	5 + 00W
Remarks	Step ~75m east of DDH-98-3 and ACE-02-03, test for mineralized horizon, felsite.	Elevation	1025 m

Azimuth	180
Dip	-45
Length	117.35 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
0.00	7.62	**Casing**	Cased through road fill and glacial till, no recovery.														
7.62	25.30	**QMBS**	**Quartz-Muscovite-Biotite Schist**; pale grey, slightly greenish, siliceous and schistose, weakly calcareous; consistent texture, well-foliated. Quartz-musc dominate with 5% brownish spotted biotite, minor chlorite. S2 @ 75-85 to c.a. Weak sulphides, weakly magnetic - pyrrhotite (po) associated with dark biotite spots. Very competent, weakly fractured, 100% recovery.	**2**	**1**												
			7.6-10.8: Darker, calcareous & chloritic sections in spotted quartz musc schist.														
			10.8-14.6: Well-fractured pale spotted schist, "poker chip" sections.														
			10.9: S2 @ 85 to c.a.														
			10.67-13.41: 70-75% recovery.														
			12.3-12.4: Weakly gougy fracture zone.														
			13.41-15.54: 95-100% recovery.														
			15.4-15.9: Marble; dark grey, dirty chloritic schist.														
			17.5-25.3: Very weakly fractured, mainly good spotted schist, more chloritic and some calcareous layers.										**17.5**	**19.0**	**8199**		
25.30	28.10	**QMCS**	**Quartz-Muscovite-Chlorite Schist;** medium greenish grey, fine-grained and weakly schistose, with grey marble layers and lenses. Chlorite occurs as green wispy layers. Gradational with above unit and lower marble unit. Slight increase in po + py, mainly as fine diseeminations along S2.	**2-3**	**1**								**25.5**	**27.0**	**8200**		
			25.3: 5cm marble layer marks contact @ 85 to c.a.														
28.10	33.80	**Marb**	**Marble;** medium grey, well-layered but strongly contorted, moderate to strong reaction with 10% HCl, dirty, micaceous marl. Weakly to moderately fractured, Po-py weaker but finely disseminated in less limy sections. Many irregular white quartz veinlets or segregations, usually not mineralized.	**1-2**	**1**								**32.0**	**33.8**	**8201**		
			29.0: Limonitic fractures.														
			29.4: 8cm white quartz vein along S2 @ 80-90 to c.a. 5% po, 2% py.														
			31.5: Irregular white to grey quartz vein, 5-8cm thick.														
			31.7: 9cm white quartz vein, 75-90 to c.a.														
			31.9: 10cm fractured white quartz vein.														

Diamond Drill Log | DDH No. ACE-02-04 | Date June 4,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number
33.8	42.5	Fels	**Felsite**; pale greenish, very fine-grained, siliceous-looking, and hard. Grades from QMCS to micaceous (musc, biotite) felsite over top 40cm, several grey banded marble intervals, as noted. Weak mineralization for felsite interval, consisting mainly of pyrite, very weakly magnetic.	<1	3-4								33.8	36.0	8202
													36.0	38.0	8203
													38.0	40.0	8204
			35.7-36.1: Well-fractured, moderately limonitic.										40.0	42.5	8205
			36.1-36.4: Marble; pale cream to medium grey, well-layered @ 80 to c.a.												
			38.0-38.2: Well-fractured, moderately limonitic.												
			38.2-38.3: Marble; as above, strongly calcitic.												
			39.4-39.5: Marble; as above, 14cm thick interval @ 80 to c.a.												
			42.5-42.6: Marble; medium-grey, calcitic, 12cm thick @ 80 to c.a.												
42.5	43.5	Arg	**Argillite / Fault**; dark grey to black, graphitic with thin white quartz layers, very contorted, intensely deformed. Strong graphitic gouge at top of unit, likely lost up to 40cm of core. Quite pyritic but very weakly magnetic.	<1	3-4								42.5	43.5	8206
			42.67-44.20: 70-75% recovery.												
			43.5: Undulating polished graphite slip marks sharp lower contact @ 60 to c.a.												
43.5	45.7	Fels	**Felsite**; pale greenish, as before (33.8-42.7m).	<<1	5-7								43.5	43.75	8207
			43.5-43.7: 20% pyrite, medium-grained in stockwork in moderately calcareous section.		20								43.75	45.7	8208
			45.0: Patches of fine-grained galena and possibly fine honey sphalerite associated with pyrite; forms wispy bands along folded layering.				1	2							
45.7	46.2	Arg	**Argillite / Fault**; same as above (42.7-43.5m), more calcareous. Strong fine-grained pyrite along S2, several planar graphitic slips.		7								45.7	46.2	8209
			45.7: 2cm black gouge, moderate graphite and pyrite.												
			46.2: Polished graphite slip along lower contact @ 55 to c.a., close to planar.												
46.2	53.8	Fels	**Felsite**; pale greenish, as before (33.8-42.7m), pyrite is stronger along weak foliation and disseminated throughout. Not magnetic, possibly traces of chalcopyrite. Very consistent pale grey, siliceous-looking with only minor pale green muscovite. Hard and weakly fractured.	<<1	7								46.2	48.0	8210
													48.0	50.0	8211
													50.0	52.0	8212
													52.0	53.8	8213
53.8	54.3	Arg	**Argillite / Fault**; same as above (42.7-43.5m), not calcareous. Strong pyrite along S2, several planar graphitic slips, minor quartz. Well-fractured.		5										
			53.8: upper slip, weakly graphitic, @ 65 to c.a.												
			53.34-54.25: 85-90% recovery, losses in argillite unit.												
			54.25: Several graphitic slips in rubble at lower contact.												
													54.25	54.6	8214

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
54.3	55.7	Fels	**Felsite;** darker grey due to increased sulphides and chlorite, very contorted, micaceous felsite unit, quite strongly mineralized throughout with mainly pyrite and increasing fine-grained pyrrhotite.										54.6	55.7	8215		
		Arg	**54.25-54.6: Semi-massive Suphides;** mainly pyrite, very weakly magnetic. Pyrite is forms bands up to 3cm thick, swirled, locally disseminated.	<1	20												
			54.6-54.8: Pale olive green, quartz-muscovite schist, likely a metadyke or sill. Fine grained with S2 @ 80 to c.a., may cut sulphide horizon at low angle.	-	-												
			54.8-55.7: Wispy bands of pyrite and po, moderately magnetic, locally > 10%, with bands of quartz, chlorite, and pale pinkish garnets. **Semi-Massive Sulphides** from 55.5-55.7m, strongly magnetic.	3-5	3-5												
55.7	56.6	Arg	**Argillite;** dark grey, moderately-layered.with minor contortions; numerous grey siliceous layers, 1-2mm thick, no calcareous bands, occasional wispy to disseminated pyrite in discontinuous bands.	<1	2								55.7	56.6	8216		
			55.7-55.8: Folded with intervening grey quartz veins, occ grains of sphene(?).														
			56.4: S2 @ 75 to c.a.; dark fine-grained silvery mineral - ilmenite-rutile(?).														
			56.6: Contact is weakly gougy, clay and chips in box.														
56.6	58.7	Fels	**Felsite;** pale-grey, fine-grained, lacking micas, with apparent fractured-healed (cataclastic) texture. Weak wispy pyrite and very weakly magnetic at start, building to near semi-massive zone. Weakly fractured.	5	3								56.6	58.7	8217		
			56.6: 1mm orange-brown limonite-ankerite spots.														
			57.3: Increasing fine-grained disseminated po, becoming net-textured, assoc py.														
			57.5-57.7: 15-20% po + py, good net texture; po > py.	15	5												
			58.3-58.5: 10-15% po + py, good net texture; po > py.	10	4												
			58.5-58.8: Strongly fractured.														
58.7	60.3	Arg	**Argillite;** dark grey - darker than above argillite, very well-layered, not contorted, much less siliceous; limestone layer 58.8-58.9m, otherwise weakly calcareous. Occasional wispy to disseminated pyrite in discontinuous bands.	<1	2												
			58.9: S2 @ 65 to c.a., very planar.														
			60.0: Some contortion of layers, weak fold; pyritic.														
			60.3: S2 & contact @ 70 to c.a., planar.														
60.3	64.9	Fels	**Felsite;** pale grey, fine-grained, locally layered with generally weak muscovite and chlorite, some mottling with carbonate, and cataclastic texture. Up to 10% po + py, mainly fine-grained and disseminated, often together. Moderately magnetic. Hard and weakly fractured.	5-7	2-3								60.3	62.6	8218		
			60.7: Tan-coloured patches and bands, hard, possibly K-feldspar(?).														

Diam d Drill Log   DDH No. ACE-02-04   Date June 4,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
			60.9: 5cm section of well-layered, pale to medium-grey calcitic limestone.														
			61.6-61.7: Blebby bronze po, strongly magnetic, lesser py.														
			62.0-62.3: Cataclastic texture - healed bx.														
			64.5: 2cm siliceous layer with 50% po + py, net-textured, along S2, layering.														
			64.8: S2 @ 80 to c.a., wavy, not planar, micaceous.														
64.9	65.9	Marb	**Marble;** dark grey, well-layered with fine pale grey calcitic and darker grey layers, locally quite contorted. Po + py continue fine-grained, somewhat weaker than in felsite. Unit is moderately fractured.														
65.9	67.2	Fels	**Felsite;** pale grey, fine-grained, locally layered with generally weak muscovite and chlorite, some mottling with carbonate, and cataclastic texture. Up to 10% po + py, mainly fine-grained and disseminated, often together. Moderately magnetic. Hard and weakly fractured.	5-7	2-3								65.9	67.2	8219		
			67.8: Becoming increasing calcareous, layers of strong reaction to HCl.														
67.2	76.0	CS	**Calc-Silicate Schist;** grades from fine-grained felsite to slightly coarser grained, locally calcitic, quite micaceous, mainly muscovite, siliceous with occasional quartz eyes Some bands of felsite remain. Continues pale grey to greenish and brownish in biotite sections, very similar colour and texture and fracture characteristics to felsite. Weaker po + py, finely disseminated along foliation. Very weakly fractured until lower contact. Could be an alteration.	2-3	2-3								67.2	69.0	8220		
			68.5: S2 @ 75 to c.a.														
			72.9: S2 @ 70 to c.a.; curved face, minor contortions.														
			75.3-76.2: <20% recovery around a quartz vein, possible fault at "contact".														
76.0	82.1	Fels	**Felsite;** micaceous and weakly calcareous, diminishing downhole; increasing po + py. Fine-grained matrix returning with occasional calcareous bands, generally more cataclastic. Becomes increasingly calcareous, much as above. Very hard and weakly fractured.										76.0	78.0	8221		
													78.0	80.0	8222		
													80.0	82.1	8223		
			77.7-78.4: Sharp increase in fine-grained po + py. Blebby py ~78.1m, less po.	2	10												
			77.8-79.5: Strong siliceous-looking felsite, no calcareous bands.														
			79.5-82.1: Several calcareous bands and veinlets; 10% po > py.														
			80.9-81.1: Quartz vein,irregular, over 50% very fine-grained brown patches, possibly tourmaline, with blebby po and py. Vein appears to be @ 20 to c.a.														
82.1	82.9	Marb	**Marble;** top marked by very calcitic, dark grey banded limestone, continues less calcitic and more siliceous. Moderate reaction with HCl.														
			82.4-82.9: Pale greenish, carbonate patches in matrix, po > py along bands.	8	2												
			82.6: 3-4cm white quartz vein, lesser carbonate, good patchy po.	10													

Diam d Drill Log DDH No. ACE-02-04 Date June 4,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number
82.9	84.1	Fels	**Felsite;** much as above, continuing carbonate patches and layers, locally cataclastic; brownish colour due to fine biotite - less than 5% micas. Variable po, py.	3-4	3-4								82.9	84.1	8224
84.1	84.5	Marb	**Marble;** mainly dark grey, well-layered, marbly texture. Weaker po + py. Upper contact @ 60, lower @ 45 to c.a.	<1	1-2										
84.5	88.4	Fels	**Felsite;** much as above, continuing carbonate patches and layers, locally cataclastic; brownish colour due to fine biotite - less than 5% micas. Variable po & py, much more pyritic. Weakly fractured, competent.	<1	5-7								84.5	86.5	8225
			84.5-87.8: Brown, biotite-rich, moderately calcareous, similar to section of calc-silicate schist. Biotite occurs as fine bands in fine-grained "felsite" matrix; carbonate occurs along these bands and as discrete strongly calcitic bands. Up to 10% py, weak po.	<1	7										
			87.8-88.4: Sharp contact with pale green, siliceous and weakly layered with drop in biotite and pyrite and increase in muscovite and po. Contact @ 55 to c.a. Transitional with quartz-muscovite-chlorite schist below.	3-4	1										
88.4	91.1	QMCS	**Quartz-Muscovite-Chlorite Schist;** pale green to dark greenish grey, moderately to well-layered, largely contorted. Occasional limy layers, grades to darker grey argillite. Sharp drop in po + py. Weakly fractured.	<1	<1										
91.1	92.1	Fels	**Felsite;** pale greenish grey, well-layered 1-3mm, more micaceous - esp musc- than most felsite intersections, not calcareous, quite contorted to cataclastic. Modest po > py. Hard and weakly fractured.	3-4	1								91.1	92.1	8226
92.1	94.5	Arg	**Argillite;** dark siliceous non-calcareous schist with 40% QMCS. Strongly contorted layering, fine, needle-like crystals, up to 3mm long scattered throughout. Weak sulphides, mainly pyrite, very weakly magnetic. Moderately fractured.	1	3-4								92.1	94.5	8227
94.5	102.4	QMCS	**Quartz-Muscovite-Chlorite Schist;** pale green to dark greenish grey, moderately to well-layered, largely contorted, as above. Occasional limy and darker grey argillite layers. Fine needle-like crystals continue, weaker. Continued weak po + py. Weakly to moderately fractured.	2	<1										
			94.6: S2 @ 70 to c.a; wavy.												
			101.8: S2 @ 75 to c.a.; planar.												
102.4	104.3	Fels	**Felsite;** pale green, quite micaceous, grading back into QMCS at both contacts. Well-layered with brown biotite and carbonate between felsic layers - moderately calcareous. Variable po > py. Moderately fractured near the top.	3-4	1-2								102.4	104.3	8228

Diam d Drill Log DDH No. ACE-02-04 Date June 4, 2

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
			102.7-102.8: Weakly layered, vuggy texture, no increase in sulphides, minor Cr-mica. Note 1-2mm crosscutting pyrite veinlets.														
			103.9-104.3: Well-layered.														
			104.2: S2 @ 72 to c.a.; planar.														
104.3	117.35	**QMCS**	**<u>Quartz-Muscovite-Chlorite Schist</u>**; pale to medium greenish grey, non-calcareous, moderately to well-layered, weakly contorted. Into monotonous sequence typical of the footwall. Fine needle-like crystals not apparent. Continued weak po + py. Weakly to moderately fractured.	**2**	**<1**								**110.0**	**111.5**	**8229**		
			104.3-105.5: Well-fractured, close to "poker chips"; S2 @ 80-90 to c.a.														
			104.5: 3-8cm thick white quartz vein with tan carbonate patches (10-20%), no po.														
			105.5-109.6: Moderately fractured.														
			109.7: S2 @ 78 to c.a., planar, weakly crenulated.														
			116.3-117.35: Strongly fractured, close to "poker chips" S2 @ 80-90 to c.a. Lost ~ 30cm of core.														
			END OF HOLE = 117.35 metres														

Diamond Drill Log

DDH No. **ACE-02-05**

Company:	**Barker Minerals Ltd**	Date Collared	June 3, 2002
Project:	**Frank Creek**	Date Completed	June 4, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	5 + 00N
Date:	June 6, 2002	Easting	6 + 25W
Remarks	Vertical hole to test gravity anomaly with southerly dip and coincident mag high.	Elevation	1027 m

Azimuth	n/a
Dip	-90
Length	67.97 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
0.00	7.62	**Casing**	Cased through road fill and glacial till, no recovery; bedrock likely ~6.5m.														
7.62	22.20	**Fels**	**Felsite;** pale grey with weakly brownish and greenish sections corresponding to increased biotite and muscovite, respectively. Most of the unit consists of very fine grained quartz or plagioclase with highly variable micas and up to 10% po + py, is well-layered and weakly to strongly calcareous with marble layers. Most of po - py is fine-grained and disseminated, aligned with micas along foliation, some py is secondary, coarser or occurs in crosscutting veinlets. Quartz veining is quite common. Interval is weakly fractured, more in marble layers.	**3-5**	**3-5**								**7.6**	**10.0**	**8230**		
													10.0	**12.0**	**8231**		
			7.6-8.5: Cataclastic texture, no layering, stronger po > py. S2 @ 60 to c.a.														
			8.5-11.7: Well-layered, brownish biotite and increasing carbonate in bands. Po - py are disseminated throughout though variable.														
			11.7-11.8: Finely-layered, dark grey quartz-musc-chl schist, strongly folded layers.														
			11.8-13.1: Pale greenish, finely-layered felsite, weakly calcareous.														
			13.1: Layering @ 70 to c.a.														
			13.1-13.3: Marble; finely-layered, medium-grey, calcitic.														
			13.3-14.1: Calcareous, pale green section.														
			14.1-16.7: Pale greenish, finely layered, weakly calcareous.														
			15.5: S2, layering @ 60 to c.a.														
			16.7-17.5: White quartz vein, continuous, with 5% tan carbonate patches, minor muscovite and weak po only along selvages. Vein appears conformable.										**16.7**	**17.5**	**8232**		
			17.5-19.2: Pale green, contorted schistose felsite cut by irregular white quartz veining. Finer po > py.														
			19.2-21.0: More thickly layered, S2 @ 55-65 to c.a., not strongly mineralized.														
			21.0-21.1: Strong fracture zone.														
			21.2-22.2: Increasing musc-rich layers, grading to schist.														
22.20	24.70	**QMS**	**Quartz-Muscovite Schist;** pale green, very siliceous, finely-layered, not calcareous, weakly mineralized, weakly magnetic. Moderately fractured.														
			22.2: S2 @ 65 to c.a., not contorted.														
			24.6: S2 @ 65 to c.a.; not contorted.														
24.70	29.60	**Fels**	**Felsite;** much as before, more cataclastic with several white quartz veins also disrupted. Moderate po + py, weak to moderate magnetism.	**3-4**	**2-3**								**24.7**	**27.0**	**8233**		
													27.0	**29.6**	**8234**		

Diam d Drill Log DDH No. **ACE-02-05** Date June 6,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
			25.8-26.2: Well fractured white quartz vein, tectonically disrupted, locally vuggy with fine-grained po along selvage, coarse py > po.														
			26.3-26.9: Similar disrupted white quartz veining, less sulphide, locally vuggy.														
			26.9-29.6: More uniform brownish, non-calcareous more weakly mineralized.														
29.6	36.9	QMS	**Quartz-Muscovite Schist**; soft, pale green, less siliceous, more massive-looking, not calcareous, weak po + py, weakly magnetic. Weakly fractured.	1	<1												
			30.7: S2 @ 40-45 to c.a., planar.														
			34.1: S2 @ 45-50 to c.a., planar.														
			34.8: Layering becoming quite contorted, erratic folding.														
			35.7-36.9: Moderately fractured.														
36.9	42.5	Arg	**Argillite**; dark grey, fine-grained and well-layered but strongly contorted and folded; quite siliceous, 20% pale grey calcareous layers, not significantly carbonaceous. Sharp increase in secondary coarser-grained pyrite, minor fine-grained lenticular po. Occasional needle-like white crystals. Uniform interval, weakly fractured.	1	3												
42.5	48.0	QMS	**Quartz-Muscovite Schist**; soft, pale green, less siliceous, still strongly contorted, weakly calcareous along thin 1-3mm pale grey layers, weak po + py, weakly magnetic. Weakly fractured.	1	<1												
			42.5: Gradational upper contact into musc-rich schist.														
			44.2: S2 @ 55-60 to c.a., relatively flat.														
			48.0: Gradational lower contact, relative decrease in musc, becomes darker.														
48.0	51.3	Arg	**Argillite**; as above (36.9-42.5m), dark and very contorted.														
			48.2: S2 @ 55 to c.a														
			49.8: S2 @ 15 to c.a.														
			50.4: Tight folds suggest a antiformal closure uphole, to south.														
			50.5: S2 @ 30 to c.a.														
			51.1: Begin to see spotty to patchy ankerite alteration.														
51.3	54.5	Fels	**Felsite**; sharp contact across white, unmineralized quartz vein, may have been more musc-rich. Pale greenish to brown and mottled, hard, not well-layered but layering appears to fold around an axis at 52.2m. Py > po, weakly magnetic. Weakly fractured.	2	4												
			53.9: S2 @ 35 to c.a.														
			54.5: Lower contact @ 65 to c.a.														

Diam d Drill Log DDH No. **ACE-02-05** Date June 6,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
54.5	56.4	Arg	**Argillite;** dark grey, well-layered, as before (36.9-42.5m), but much less contorted and folded, more siliceous layers. Still not strongly carbonaceous. Significantly more po & py, esp along layers. Weakly fractured.	3-5	5												
			55.2: S2 @ 55 to c.a., planar, polished.														
56.4	57.7	Fels	**Felsite;** sharp contact at end of run. Pale greenish to brown and mottled, hard, not well-layered, but locally stockwork breccia. Py > po, weakly magnetic. Moderately fractured.	2	4												
			57.1-57.4: Strongly cataclastic with well-developed pyrite stockwork, minor po. Dark hairline fractures(?) and bright green Cr-mica.														
			57.6: Cr-mica along layer associated with py.														
57.7	63.3	Arg	**Argillite;** dark grey, well-layered, as before (36.9-42.5m), again quite contorted and folded, more siliceous layers. Somewhat more carbonaceous, occasional graphitic slips. Significantly more po & py, esp along layers. Weakly fractured.	3-5	5												
			59.0: S2 @ 50 to c.a., wavy.														
			62.3-63.3: Moderately fractured, several curved and planar graphitic slips.														
			62.5: S2 @ 55 to c.a., planar, mod graphitic slips; 62.6: S2 @ 0 to c.a., steepens.														
63.3	63.8	Flt	**Fault;** only 12cm of soft very plastic clay to granular gouge recovered, assumed to be where core was lost.														
			62.48-65.53: 85-90% recovery.														
63.8	66.5	QMS	**Quartz-Muscovite Schist;** pale green, siliceous, locally contorted, weakly calcareous along 1cm po-rich layers, mod po + py, weakly magnetic. Weakly fractured.	2-3	2-3												
			65: 1-2cm layer of 25% po, min py + cp, in net-texture, weaker to 65.2m.														
			65.2-65.5: Felsite; slice of well-mineralized pale greenish-grey, fine-grained felsite, as before. Py - po both blebby and disseminated.	5	5												
			65.7: Layering (S2) @ 0 to c.a. over 10cm.														
			66.1-66.2: Irregular white quartz vein, rimmed by patchy tan carbonate over a total width of 10cm, @ ~ 40-50 to c.a. Sulphides limited to wall rock.														
			66.5: Lower contact @ 70 to c.a.														
66.5	67.97	Arg	**Argillite;** contorted dark banded schist, as described previously. Not calc, sharp increase in pale needle-like crystals. Moderate fine po & coarse py.	2	3-4												
			END OF HOLE = 67.97 metres														

Diamond Drill Log DDH No. **FC-02-01** Tests

Company:	**Barker Minerals Ltd**	Date Collared	May 15, 2002
Project:	**Frank Creek**	Date Completed	May 16, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	L57+95N
Date:	May 16, 2002	Easting	16+00W
Remarks	Test Trench 10 sulphide occurrence.	Elevation	1042 m

Azimuth	090
Dip	-45
Length	97.84 m
Core Size	NQ

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
0.0	6.10	**Casing**	Cased through glacial till, no recovery.														
6.10	11.20	**QSP Bx**	**Quartz-Sericite Phyllite-Breccia;** pale to medium grey, fine-grained, sericite-rich	-	3-4				0	2	3	2					
			breccia textures, with angular to subrounded clasts in phyllitic matrix. Phyllitic to weakly schistose with pyrite-rich bands, often coarse-grained and cubic - pyritic bands can crosscut foliation. Foliation (S2) is at 60 to c.a. at 6.3m. Recovery >95%, RQD ~ 50%.														
			7.5: Minor Fault; 2cm pale grey clay gouge @ 70 to c.a.														
			9.1: Minor Fault; 5cm pale grey clay gouge @ 60 to c.a.														
			9.2-10.2: Pale grey more massive section, strongly sericitic. Looks like felsic lapilli tuff unit with 0.5cm lapilli oriented along S2.														
			10.2: Weakly gougy slip at contact.														
			10.2-11.2: Breccia, likely deformational, with subrounded blocks to 3-4cm in grey phyllitic matrix; S2 does not penetrate larger clasts. Clasts show qtz-eyes --> felsic volcanic bx.														
11.20	15.90	**Arg**	**Argillite;** dark grey to black, strongly foliated and well-banded, biotite(?)-qtz-graphite phyllite; relatively soft. Strongly contorted with fine tight to isoclinal rootless folds. Siliceous bands are moderately sericitic, possibly silty interbeds or silica segregations. Pyritic with up to 10% over 20cm, 5% overall; concentrated in siliceous bands. Not magnetic, no other sulphides identified.		5				0	1	1	1					
			11.9: Finely banded, poss original layering, <1mm thick @ 80 to c.a.														
			15.0-15.2: 10-15% pyrite.										14.9	15.9	8235	60	109
15.90	18.20	**QSP-Bx**	15.9-16.0: White quartz vein, largely shattered, barren.		2-3												
			16.0-16.6: Finely laminated phyllite, bo bx textures.														
			16.6: Minor fault; 2cm grey clay gouge @ 65 to c.a. **16.6-17.2: Argillite**.														
			17.2-17.3: Fault; 13cm of pale grey clay gouge bx.														
			17.6-17.7: Argillite layer.														
18.20	20.80	**Arg**	Argillite; contact @ 80 to c.a., very sharp transition from granular-looking wacke		3-4												
													19.6	20.6	8236	79	234
			18.7; Minor fault; 1cm soft, sericitic gouge.														
			19.0-19.4: Fault; dark grey pebbly fault bx.														

Diam[illegible]d Drill Log   DDH No. **FC-02-01**   Date May 16,[illegible]2

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
20.8	24.9	QSP	**Quartz-Sericite Phyllite;** pale-greenish-grey, fine-grained, weakly phyllitic, from well-banded to breccia, as above (6.1-11.2m). Pulses of py-cp mineralization, as noted below. Competent, non-calcareous, strong sericite altn.		3	tr	tr	tr		1	3-4	1	20.6	22.0	8237	125	994
													22.0	23.5	8238	93	1725
													23.5	24.9	8239	80	1085
			20.8: Contact, S2/l0 @ 80 to c.a.														
			20.8-21.6: Finely laminated section, weakening downhole.														
			21.6-24.9: Mainly massive to brecciated section, increased sulphides.														
			22.4: 6cm bleb of 15-20% py, 2-3% galena, possible traces of sphalerite. Fine-grained, cubic pyrite, finer identifiable galena.														
			23.6: 1cm band of honey sphalerite (?) with coarse pyrite, minor galena and cp.														
			24.4: Minor fault; 1-2cm pale grey clay gouge.														
24.9	26.9	Arg	**Argillite;** dark grey, massive to finely laminated silty mudstone less phyllitic but with locally contorted S2 (25.9m). Coarse pyrite common, distributed throughout.		5	<1	tr	tr		1	2	1	24.9	26.9	8240	88	2031
26.9	29.8	QSP	**Quartz-Sericite Phyllite;** as before, 20.8-24.9m, with darker, laminated phyllite more dominant, becoming more mottled and possibly brecciated @ 28.6m.										26.9	27.3	8241	1183	2.30%
													27.3	29.1	8242	217	2061
			26.9-27.2: 5-10% pyrite, mainly coarse cubic, with 2% coarse patchy cp, 1% honey sphalerite, and <1% silvery galena.		10	2	1	<1					29.1	29.5	8243	1461	3.40%
			29.8: Minor gougy shear @ 60 to c.a.										27.30	29.10	2LF01-31	38	476
			28.4: Minor fault; 5cm pale clay-sericite gouge.										28.96	29.10	3LF01-32	2450	3.42%
			29.1-29.5: Sulphide zone with 5-10% coarse pyrite, 1-2% cp, 1% sphalerite and minor galena in siliceous grey matrix.										29.10	29.50	2LF01-44	1950	3.02%
			29.3: S2 @ 52 to c.a.														
29.8	30.8	Flt	**Fault;** pale grey gougy and brecciated grey phyllite. Top slip @ 20 to c.a., bottom contact between gravelly bx and phyllite 15-20 to c.a.										29.5	30.8	8244	57	622
30.8	47.5	Arg	**Argillite;** dark grey, fine-grained, as before. Increasing graphite (moderate) and sericite, decreased quartz makes unit softer and more fractured. Locally well-laminated, often contorted with core axis angles ranging from 60 (33.1m), to 85 (40.3m). 2% cubic pyrite, locally 5%.		2					1	1-2	2					
			30.6-36.5: Well-fractured, 95% recovery.														
			37.0-37.1: Layer of brecciated siltstone (grey phyllite) or possibly lapilli tuff.														
			42.0: 1-2cm band of coarse cubic pyrite.														
			42.8-43.1: Pale grey phyllite good breccia textures; 10% cubic pyrite.														
			43.1: Minor fault; 5-10cm of clay-ser gouge.														
			43.8-43.9: Minor fault; 10cm of dark grey, weakly graphitic gouge.														
			46.0-46.2: Grey phyllite hosting 1-2cm thick contorted grey quartz vein. A few 1mm blebs of cp noted.														

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
			46.3: Polished graphitic plane @ 40 to c.a., becoming more prevalent.														
			46.9-47.5: Well-fractured, "poker chips".														
			46.94-48.73: 45% Recovery.														
47.5	49.0	Flt	**Fault;** poor recovery in dark graphitic gouge and polished graphitic schist and bx.														
49.0	51.3	QSP	**Quartz-Sericite Phyllite;** pale greenish to grey, fine-grained but with a few small clasts evident. Strongly sheared, with fault breccia as noted. Occasional coarse pyrite.		2												
			49.0-49.5: Competent, almost gritty siltstone.														
			49.5-50.4: Moderate intensity fault; quite gougy, esp last 5cm - @ 65 to c.a.														
			50.5-50.5: Fractured white quatrz vein.														
			51.2: 1-2cm band of c-gr pyrite, 25%.														
51.3	59.4	Arg	**Argillite;** dark grey, finely laminated with 25% pale grey siliceous bands, ~0.5mm thick. Dark polished graphitic planes, mod graphitic. Highly contorted and folded pale grey layers over top 80cm, then fairly uniform with S2 @ 45 to c.a. (52.5m). Pyrite, both fine and coarse, occurs preferentially in the pale siliceous bands.		3-4				0	1	2	2					
			Relatively competent, 100% recovery.														
			55.2: Weak shear zone, 5-10cm thick.														
59.4	60.1	Flt	**Fault;** dark, mainly graphitic mushy gouge; 5cm matrix-supported fault bx at top, planar slip @ 50 to c.a. Pyritic black sand over last 30cm. Recovery ~50-60%.														
60.1	68.2	QSP	**Quartz-Sericite Phyllite;** pale grey phyllite, locally with clasts of argillite near top, extensively sheared and healed, becoming spotted, as described below. Coarse pyrite evident but not strong. Increased tan cb-altn, minor Cr-mica, esp 63.3-65.5m.										60.1	61.5	8245	40	1379
			60.1-61.5: Gritty appearance, with a series of healed S2-parallel fractures.														
			61.5-61.7: Fault; grey gouge, upper slip @ 50 to c.a. (parallel to S2).														
			61.7-62.3: Well-fractured, locally gougy slips.														
			62.9-63.3: Fault; gougy fault breccia, clasts to 3cm.														
			63.9-64.0 Fault; gouge.														
			64.3: Contact between argillite layer and grey gritty phyllite @ 55 to c.a.														
			65.5-68.1: 15-20% pale grey "spots", 2-3mm diam, dolomite or ankerite alteration?														
			67.0-67.3: Well-fractured, weak shear zone.														
			67.3-67.6: Well-layered or laminated, with dark argillite @ 65 to c.a.														
			67.3: 1-2cm gougy seam.														
			68.1-68.2: 7-8cm quartz vein or sweat @ 60 to c.a., in argillite.														

Diamond Drill Log DDH No. **FC-02-01** Date May 16, .

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
68.2	71.3	Flt	**Fault;** 3cm of soft, plastic clay @ 80 to c.a., marks top of fault zone. Highly fractured throughout, often as a good breccia.														
			68.3-69.7: Weakly "cemented" fault breccia with poorly sorted, angular to subrounded clasts to 5cm, in a dark, weakly graphitic matrix. Recovery ~75%.														
			69.7-70.5: Cracked grey siliceous unit.														
			70.6: Pale olive green clay gouge.														
			70.7-71.3: Mainly grey phyylite and quartz rubble; ~20% recovery - matrix washed away?														
71.3	74.7	QSP	**Quartz-Sericite Phyllite;** variable unit with "spotted" texture, argillite layers, and breccia related to faulting on both sides of the interval. Continues non-calcareous despite faulting. Fractured quartz veining evident. Pyrite weaker.		1-2				1	1	3	3					
			72.1-72.2: Significant quartz veins or sweats, white and barren, minor sericite.														
			72.6-72.9: Well-layered, as before. S2 @ 75 to c.a.														
			73.1-73.2: Pale greenish to cream carbonate alteration, with quartz eyes, 1mm spots of Cr-mica, adjacent to brecciated quartz vein. Qtz eyes preserved.														
			73.2-73.5: Coarsely brecciated quartz vein, white to cloudy grey, barren.														
74.7	77.2	Flt	**Fault;** increasing breccia intensity, with matrix becoming very soft clay, with fewer and smaller hard grey clasts. Weakly cemented, locally rubbly.						1	1	4	5					
77.2	81.5	QSP	Pale to med grey, locally well-layered, contorted and brecciated, grading to arg.														
			77.3-77.4: Pale greenish lapilli-shaped clasts in dark grey, graphitic matrix. Clasts up to 4 x 1.5cm, wide range in size and shape. S2 @ 40 to c.a.														
			79.7: Open fold, layering wraps but no penetrative axial planar foliation														
			79.8: 1cm grey gouge.														
			79.9: Becoming increasingly "spotted" and gritty, a few cm-size pyrite cubes.														
			81.4: 5cm wispy bandof tan carb altn, several good flecks of Cr-mica.														
81.5	82.7	Flt	**Fault;** medium grey gougy fault breccia with grey phyllite clasts supported in soft grey clay. Both contacts @ 50-55 to c.a., undulating slip planes.														
82.7	87.2	QSP	**Quartz-Sericite Phyllite;** pale grey to pale olive green, fine-grained to weakly gritty, quite uniform. Moderately fractured, but not strongly sheared or brecciated, as before. Locally spotted texture, coincident with greenish sericite. Less pyrite, occ 1cm cubes. Incr tan carb altn, max at l.c. @ 55 to c.a.														
			85.8: S2 @ 70 to c.a.														
87.2	97.8	QSCP	**Quartz-Sericite-Chlorite Phyllite;** dark grey, quartz-rich, non-graphitic, fine-grained, finely foliated phyllite, often gritty and spotted, and weakly banded. Little shearing, core quite competent and good recovery. Upper contact @ 55 to c.a., pyrite cubes more common in top 30cm.										89.1	90.5	8246	14	45

Diam[illegible]d Drill Log DDH No. **FC-02-01** Date May 16, [illegible]

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number		
			87.8: 3-4cm thick quartz vein @ 65 to c.a., gougy 1cm shear on fw of vein.														
			88.0-88.1: Weak shear zone with white quartz veining, gougy in centre @ 35 to c.a.														
			88.1-92.2: Weakly foliated, dark grey gritty interval, more massive-looking.														
			92.6: Minor shear @ 25 to c.a.														
			93.1-93.6: Dark grey gritty unit. 93.1: S2 @ 75 to c.a.														
			93.6: Quartz vein-minor shear zone, 5-7cm thick.														
			93.7-96.3: Well layered, 1-2mm, contorted section, very soft, increasing shearing @ 95.0-95.3 and 95.6-95.9m.														
			96.3-96.9: Pale olive green alteration of darker gritty unit.														
			96.9-97.4: Well-layered soft, dark phyllite, as before.														
			97.4-97.8: Several cm thick white quartz veinlets, generally conformable with S2 @ 65 to c.a.														
			END OF HOLE = 97.84 metres														

Diamond Drill Log

DDH No. FC-02-02

Company:	**Barker Minerals Ltd**	Date Collared	May 16, 2002
Project:	**Frank Creek**	Date Completed	May 17, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	57+20 N
Date:	May 17, 2002	Easting	12+80 W
Remarks	Lower switchback, D Road; Conductor B	Elevation	1077 m

Azimuth	090
Dip	-45
Length	144.78 m
Core Size	NQ

Tests

Depth	Dip
n/a	

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
0	10.7	Casing	Cased through glacial till, no recovery.														
10.7	28.2	QSP	**Quartz-Sericite Phyllite;** pale grey to weakly greenish, fine-grained to locally gritty siltstone to sandstone/phyllite. Interval is weakly to moderately fractured, with distinctive orange Fe-carbonate alteration along fractures and penetrating phyllite at the top and bottom. Alteration is strong rust colour and reacts strongly with HCl. Minor irregular white and grey quartz veining. Well-behaved main foliation, not crenulated.														
			10.7-15.0: Strong Fe-carb alteration, intense along fractures, penetrating up to 3cm into wall rock. Away from the fractures, Fe-carb forms spots becoming less dense further from the fracture. Some sections up to 20cm thick consist of Fe-carb, with in-situ clasts of grey phyllite.														
			15.0-25.3: Fe-carbonate alteration is restricted to rusty fractures - generally weakly fractured. No spotted Fe-carb in wall rock.										**16.7**	**18.2**	**8247**	**64**	**71**
			15.0: S2 @ 70 to c.a. 22.9: S2 @ 75 to c.a.														
			25.3-28.2: Strong Fe-carb alteration, likely related to strong graphitic fault zone below. As before, alteration is centred on large fractures, penetrating into wall up to 7cm. Calcite veining in some fractures, very strong reaction to HCl.														
			26.2: Fe-carb on fracture @ 40 to c.a.														
			26.4: Calcite-Fe-carb vein, 5-7mm thick @ 30 to c.a.														
			26.6: S2 @ 75 to c.a.														
			26.8: Calcite-Fe-carb vein, 5-6mm thick @ 25 to c.a.										**26.5**	**28.0**	**8248**	**35**	**57**
			28.0-28.2: Last of Fe-carb altered phyllite, becomes rubbly.														
			28.04-29.26: Recovery = 20%, 15cm from 28.04 and 10cm above 29.26m.														
28.2	32.2	Arg	**Argillite;** dark grey to black, local fault breccia10-15% white and grey quartz veining; very well-fractured and sheared; not strongly graphitic, more chloritic. (QCSP - Quartz-Chlorite-Sericite Phyllite).														
			28.2-29.1: Not recovered, assumed to be soft fault gouge, likely graphitic.														
			29.1-29.6: Fine, dark grey weakly graphitic micaceous phyllite, py cubes to 1cm.														
			29.6-30.2: Well-fractured, 25-30% white quartz veining, in dark phyllite. Increase in frequency of coarse pyrite cubes, some loose in box.														
			30.2-31.1: Contorted S2, folded, not well-banded. S2 @ 45 to c.a.														
			31.1-32.0: Rubbly, again lots of brittle white quartz vein material.														

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
32.2	36.3	QSCP	**Quartz-Sericite-Chlorite Phyllite;** medium to dark grey phyllite, more gritty than above unit. Weakly foliated, locally massive. Quartz veining and coarse pyrite cubes still common.															
			35.0: S2 @ 60 to c.a.															
			33.8-35.5: Deformation bx, elongate clasts of pale grey siltstone in black mudstone matrix, clast & matrix-supported. Bx then contorted.															
36.3	38.7	QV/Flt	**Quartz Vein in Fault;** interval consists of thick white quartz veins at both contacts, bracketing a zone of highly silicified pale greenish phyllite.											36.3	37.3	8249	84	46
			36.3-37.3: Mainly white quartz vein with bits of phyllite between.															
			37.3-37.6: Hard silicified phyllite.															
			37.6-37.8: Soft sericitic phyllite.															
			37.8-38.1: Soft pale grey fault gouge, with pieces of white quartz vein to 3cm.															
			38.1-38.7: Mainly fractured white quartz vein.															
38.7	41.2	Flt	**Fault;** mainly highly fractured and sheared quartz-graphite schist with 5-10% irregular white veining and significant gougy shears. Conductor.															
			38.7-39.6: Pale grey clay-sericite gouge with fine white quartz chunks. Less than 50% recovery.															
			39.6-41.2: Mainly black graphitic gouge and fine rubble; only 25% recovered.															
41.2	48.7	QGS (Arg)	**Quartz-Graphite Schist;** black, highly contorted and schistose quartz-graphite schist. Polished graphite planes indicate significant shearing. White quartz veining along S2, mostly 2-5mm thick. Well-fractured.															
			41.2-43.4: Soft and well-fractured, less quartz..															
			41.6: Crenulation fold axial plane (S3) @ 15 to c.a., S2 @ 20, opposite.															
			43.4-45.0: More quartz veining, harder & less fractured.															
			46.4-46.9: Cream-coloured wisps along S2 in more pale grey, more felsic-looking section. Clearly healed bx.															
			46.9-48.7: Poorly sorted deformation breccia, mainly matrix-supported, with elongate clasts oriented along S2. Clasts are oval, mainly rounded to subrounded and pale grey in a medium grey featureless **argillite** matrix.															
48.7	51.5	QSP-Bx	**Quartz-Sericite Phyllite;** pale grey to greenish deformation breccia, with several minor gougy slips. Fine scattered green Cr-mica grains ~ 50.5m.															
			49.4-49.8, 50.0-50.3: Well fractured, locally gougy.															
			51.3: S2 @ 30-35 to c.a.															
51.5	53.5	Flt	**Fault;** mainly pale grey mushy gouge. Lower slip @ 30 to c.a.															
			50.51-53.04: 0% recovery. Block: "Fault - No Core".															

Diamond Drill Log DDH No. **FC-02-02** Date May 17,

Depth (m) From	Depth (m) To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
				Alteration score 1 - 5														
53.5	61.0	QSP	**Quartz-Sericite Phyllite;** as above, pale grey phyllite with wispy brownish cream altn and occasional grains of Cr-mica. Most of interval is competent but relatively soft. Good reaction with HCL when scratched - strong carb-Cr-mica alteration. Possibly a mafic or ultramafic rock. Competent.											53.5	55.0	8250	70	67
			Quartz veins usually 2-4cm thick, cross cut steep S2, constitute approx 5-10% of unit. Barren but often vuggy in centre.															
			56.4: S2 @ 80 to c.a.															
			59.5-60.3: Fault; minor rubbly shear zone.															
61.0	61.8	Flt	**Fault;** initially, gougy and rubbly zone consisting of carbonate altered grey phyllite, leading to strongly gougy section and related fault breccias.															
			59.5-60.3: Gougy frcture zone, rubbly at top.															
			60.3-61.0: Competent phyllite, increasing tectonic bx becoming gougy.															
			61.0-61.8: Pale grey, calcareous clay gouge.															
			61.8-63.7: Mainly coarse tectonic bx, moderately well-cemented.															
63.8	70.6	Grey	**Quartz-Sericite Phyllite;** moderately fractured, locally sheared phyllite; looks very similar to above grey phyllites but lacks significant carbonate and Cr-mica. 2-3% pyrite cubes to 5-6mm, previously lacking.															
			62.48-63.70: Only 45% recovery, likely at gougy seam `62.6.															
			63.7-63.8: Clay gouge; lower slip @ 75-80 to c.a.															
			67.4: Gougy, weakly carbonacous slip, 3-4cm gouge @ 45 to c.a.															
			68.4: 2+cm layer of **massive cubic pyrite** @ 60 to c.a., disseminated pyrite cubes more below layer.											68.3	69.1	8251	108	43
			67.6-66.7: Strong breccia clasts.															
			68.7-69.2: Strong breccia texture.															
			69.2-70.6: Well-fractured, well-banded phyllite, increasingly carbonaceous and contorted.															
			69.8: S2 @ 40 to c.a., much flatter than most of interval.															
70.6	75.6	Arg	**Argillite;** dark grey to black, finely laminated, strongly sheared with gougy sections. Becomes rubbly, poor recovery.															
			70.6: Sharp contact along S2 @ 40 to c.a.															
			71.2-75.6: Mainly graphitic rubble, only 20% recovery.															
75.6	76.0	Flt	**Fault;** intense black & grey clay gouge.															
76.0	84.1	Arg	**Argillite;** as before, dark grey to black, very fine grained and locally massive-looking. Soft and highly fractured, sometimes poker chips.															
			76.2-77.8: Rubbly, not obviously sheared, close to 100% recovery.															
			76.6-76.7: Bands of quartz veinlets.															

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
			81.6-84.1: Rubbly quartz-graphite phyllite, not gougy, 90% recovery.															
84.1	95.2	QSP-Bx	**Quartz-Sericite-Phyllite;** mainly tectonic breccia with pale grey clasts, subrounded and oval, very poorly sorted, in a darker grey, very fine-grained mudstone matrix. Some shearing evident parallel to S2; S2 generally quite steep to c.a. Relatively competent, weakly to moderately fractured. Not strongly pyritic.															
			84.1-84.4: Finer bx, well-fractured to rubbly.															
			84.4- 85.6: Coarse, poorly sorted tectonic bx.															
			85.6-85.7: Soft rubbly gouge.															
			85.9-86.2: Fault; grey clay gouge and white quartz vein fragments.															
			88.3: Clay gouge seam <1cm thick @ 15 to c.a.															
			90.0: 5cm pale grey clay gouge.															
			90.0-91.4: Excellent matrix-supported, poorly-sorted tectonic bx.															
			93.0-93.6: Pale grey **gritty unit**, similar to quartz eye unit described below. Weakly fractured, <1% pyrite.															
			94.2-95.2: Strong bx, as 90.0-91.4m.															
95.2	101.0	QSP Grit	***Quartz-Sericite Phyllite;*** *pale grey, fine to medium grained, siliceous with* clear glassy quartz eyes throughout, at times >25% of core. Few micas, weak S2. Hard and very competent, occ pyrite cubes. Possible felsic volcanic, but variable qtz-eye size & shape suggests clastic origin.											96.0	97.5	8252	10	24
			95.2: Upper contact is sharp, wavy @ 60 to c.a.															
			95.6-95.8: Grey phyllite with occasional grains of Cr-mica.															
			98.7-99.0: Darker grey, brecciated, silt-mudstone, as before.															
			99.7-99.9: Darker breccia section, mudstone matrix.															
101.0	114.3	QSP-Bx	**Quartz-Sericite Phyllite;** mainly tectonic breccia, with poorly sorted oval to subrounded grey silty clasts in a darker grey, fine-grained matrix. Pyrite cubes scattered throughout, <1%. Competent with occasional gougy shears.															
			102.6-103.0: Grit section, as above, fewer quartz eyes and weakly brecciated, with lapilli-like clasts in hairline matrix.															
			107.4-107.6: Grit, as above.															
			108.0: Clasts of tan-altered grey phyllite/siltstone in dark grey matrix suggests that breccia is post-deposition; ie. tectonic.															
			107.6-112.3: Strong tectonic bx, as before.															
			112.3-112.8: Grit unit.															
			112.8-114.3: Very strong breccia, more variety of clast types, shapes and sizes, matrix-supported.															
			113.0: S2 @ 75 to c.a.															

Diam d Drill Log DDH No. **FC-02-02** Date May 17,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
114.3	120.9	FAULT	**Fault;** black weakly graphitic and grey clay gouge. A few short fault breccia intervals through the clay. Recovery is between 95-100%.	2-3														
			115.6: Several cm-size pyrite cubes.															
120.9	144.8	QSP-Bx	**Quartz-Sericite Phyllite;** mixed unit, pale to dark grey, fine-grained phyllite, likely silty mudstone, much of it tectonically brecciated and healed during deformation. Also, late gouge shearing and faulting common. Pyrite cubes continue but <1%. Strong tan carb altn and spotted ankerite.															
			120.9-133.5: Mainly finely laminated fine sediments, less bx.															
			122.1-122.4: Moderately gougy shear zone.															
			122.8: Gouge-bx seam @ 20 to c.a.															
			123.1: Approx 10cm white quartz vein.															
			123.2-123.4: Fault; soft, pale grey gouge.															
			123.9: Gougy slip, 2cm thick, @ 20-30 to c.a.															
			124.2-124.5: Pale greenish, finely laminated, weakly calcareous phyllite, Fe-carb alteration. S2 @ 75-80 to c.a.															
			125.5-126.0: Grit unit, slightly softer and more sericitic.															
			126.5-127.9: Pale greenish, finely layered, quite calcareous - reacts with cold HCl - intense carb altn. S2 @ 85 to c.a.											126.5	127.8	8253	76	79
			128.6-128.8: Weak, gougy shear zone.															
			129.5-129.8: Fault; gougy shear zone with strongly polished graphitic slips over the last 10cm. S2 @ 20 to c.a.															
			129.8-131.1: Grit unit, weakly sheared throughout.															
			132.3-132.6: Fault; minor, gougy shear zone.															
			132.6-137.2: Mainly coarse tectonic bx, clasts of carb altered phyllite and grit.															
			136.8: S2 @ 80 to c.a.															
			137.2-137.5: Fault; minor, gougy shear, upper slip @ 45 to c.a., cuts S2.															
			137.7-138.0: Layering and S2 form open fold.															
			138.2, 138.6, 138.7: 1cm layers of tan carbonate altn with blebby green Cr-mica, along S2 @ 90 to c.a. Coincident pyrite cubes @ 138.6m.															
			139.0: 3cm wispy tan carb altn, with minor Cr mica, S2 @ 80 to c.a.															
			138.0-139.0: Mainly grit unit.															
			139.0-141.9: Finely laminated phyllite with faint "spots" overprinted on foliation. Spots 2-5mm, appear to be sericite.															
			141.9-142.2: Weak tectonic breccia.															
			142.2-142.5: Fine grit unit; S2 @ 80 to c.a.															
			142.5-144.78: Finely laminated phyllite, as above (139.0-141.9m).															
			END OF HOLE = 144.78 metres															

Diamond Drill Log

DDH No. FC-02-03

Company:	**Barker Minerals Ltd**	Date Collared	May 18, 2002
Project:	**Frank Creek**	Date Completed	May 20, 2002
Logger:	Christopher J.Wild. P.Eng.	Northing	54+30 N
Date:	May 21, 2002	Easting	13+47 W
Remarks	Drill east under Frank Creek Showing.	Elevation	1140 m

Azimuth	090
Dip	-45
Length	118.87 m
Core Size	NQ

Tests

Depth	Dip
n/a	

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
0.0	5.2	Casing	Cased through glacial till, no recovery.														
5.2	14.1	QSP	**Quartz-Sericite Phyllite;** pale grey, fine to medium-grained with somewhat gritty		1						3	1	8.6	9.6	8254	11	45
			Weak carb altn. Less than 1% pyrite, mainly as coarse cubes. Competent, weakly fractured, >95% recovery.														
			5.18-5.35: Weakly oxidized, fine-grained, pale greenish sericitic phyllite.														
			5.35-6.0: Coarse-grained pea-size grit. Sericitic matrix with undulating foliation.														
			6.0-10.6: Mainly homogenous pale grey, slightly gritty, weakly layered QSP.														
			10.6-14.0: Finely interlayered, pale grey siltstone and dark grey mudstone, typically 2-5mm thick. Pale cream wispy carbonate (sericite) alteration.														
			10.7: S2 @ 70 to c.a.														
			13.2-13.2: Blue quartz vein or sweat with 5% pyrite.														
			14.0: S2 @ 75 to c.a.														
14.1	14.4	QV	**Quartz Vein**, white bull quartz lacking sulphides. Highly fractured, uncertain recovery. Patchy tan carbonate.														
14.4	15.9	Flt	**Fault;** approx 20% recovery between 14.94-15.85m, block notes "gravel". Gravel includes white quartz and pale green phyllite strongly altered by Fe-carbonate + Cr-mica, and cut by narrow (1cm) quartz veins. Green phyllite marks 10cm at interval boundaries.		2												
15.9	24.2	QSCP	**Quartz-Sericite Chlorite Phyllite;** finely interlayered, pale grey silty-looking phyllite with dark grey bands of mudstone, typically 2-5mm thick. Continuing weakly pyritic, usually as cubes. A few sections of weak carbonate alteration. Interval is moderately competent and fractured with recovery estimated at 91%.		1-2												
			16.7: 5cm section of cream-coloured Fe-carbonate alteration with minor Cr-mica (~1%). S2 @ 60 to c.a.														
			17.2-18.0: Cubes of pyrite, 2-3%.														
			18.3: Wisps of sericite, Fe-carbonate with strong Cr-mica @ 75 to c.a.														
			18.7: Deformation breccia.														
			18.9-19.0: Fe-carbonate flooded along S2 @ 75 to c.a.														
			19.8-20.0: Deformation breccia.														
			20.0-24.2: Finer breccia in narrow intervals, minor wispy Fe-carb.														

Diam d Drill Log DDH No. **FC-02-03** Date May 21,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
24.2	41.1	Flt	**Fault:** Major structure, intense pale grey gouge and strongly gougy breccia.	2-3							2	4	5					
			Strongly pyritic both as coarse and fine cubes.															
			33.9-34.1: Black, weakly carbonaceous phyllite, not strongly sheared but well-fractured. Significant wispy to banded sulphides, perhaps up to 10%, mainly pyrite. Upper slip @ 40 to c.a.; recovery ~75%.															
			34.14-34.75: Recovery only 30-35%.															
			34.2-34.7: Massive Sulphides: Several cm-size pieces of pyritic massive sulphide in matrix of sulphide gouge, very little of the gouge was recovered. Sulphides are >60% of pieces, fine-grained, weakly magnetic, mostly pyrite with minor pyrrhotite, chalcopyrite, and sphalerite (?).	60	<1			?						34.2	34.7	8255	5238	3333
			34.7-41.1: Mainly pale to medium grey clay gouge and gouge breccia.															
			41.1: Very sharp planar slip between gouge and dyke @ 45 to c.a.															
41.1	41.8	AND	**Andesite Dyke:** Pale green, fine-grained, relatively fresh and undeformed. Soft yet weakly fractured. Weakly porphyritic with altered feldspar phenocrysts and quartz eyes - suggests more dacitic composition.	<1										41.1	41.8	8256	48	76
41.8	42.5	Flt	**Fault:** 10cm of clay gouge, washed out with only chips of black phyllite remaining over remainder of interval (loss of core likely here but could be in following interval).															
			41.15-42.06: Recovery ~88%.															
			42.06-43.59: Recovery ~75%.															
42.5	43.7	QSP Grit	**Quartz-Sericite Phyllite:** Pale grey quartz eye unit, medium grained, with dark phyllitic "stockwork" giving core a lapilli or breccia appearance. Unit appears to be volcaniclastic, but could be tectonically sheared dacitic volcanic. Pyrite continues mainly as cubes up to 5mm.															
			42.5: S2 plane @ 75 to c.a., top of unit.															
			43.0-43.2: Well-fractured, minor shear zone.															
			43.5-43.6: Fault; pale grey clay gouge.															
43.7	44.2	Arg	**Argillite:** Dark grey, weakly carbonaceous, tectonically brecciated interval. Sharp upper contact with clasts of "Grit" in mudstone matrix @ 75 to c.a.															
44.2	44.4	Flt	**Fault:** Relatively minor fault marked by 5-10cm of white to grey quartz vein and angular, pebbly gouge.															
44.4	45.1	AND	**Andesite dyke,** as described above, well-sheared and fractured contacts.															
			Fault at bottom, similar to 44.1-44.4m.															

Diam d Drill Log | **DDH No.** **FC-02-03** | **Date** May 21,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
45.1	50.1	QSP	**Quartz-Sericite Phyllite:** Pale grey phyllite, generally brecciated, possibly a felsic (dacitic) volcanic breccia or tuff. More siliceous with less well-developed foliation. Clasts range up to several centimetres in size, but quite variable though better sorted and less elongate than tectonic breccia. Cream to pale green "spots", up to 3mm diameter related to carb overprint.	1-2										46.5	47.7	8257	16	77
			47.3-47.7: Breccia texture with <3cm clasts of qtz eye dacite in wispy and swirled sericitic matrix.															
			45.7-45.8: Fault; minor with >5cm of clay gouge.															
			49.1-49.5: Moderately fractured with increased quartz veiing and minor gougy shears.															
			50.1: Sharp lower contact with fault.															
50.1	50.2	Flt	**Fault:** 10cm of strong, pale grey clay gouge.															
50.2	53.4	Arg	**Argillite:** Mixed dark grey and pale grey phyllites, likely alternating layers of siltstone (and grit) and mudstone. Increasing milky to granular quartz veining, somewhat contorted. Generally becomes paler and coarser down (tops?).	1-2														
53.4	54.4	QSP Grit	**Quartz-Sericite Phyllite:** Pale greenish or cream coloured, medium-grained, granular texture with well-developed qtz eyes. Strong carbonate alteration with significant 1mm grains of Cr-mica. Contacts are very sharp but irregular; upper may be alteration front, lower is lithologic contact @ 20-25 to c.a. with more intense altn, looking like a chill margin.	1-2										53.4	54.4	8258	8	137
54.4	56.2	Arg	**Argillite:** Similar to 50.2-53.4m. Mainly dark grey phyllite with up to 50% quartz-sericite phyllite, possibly silty or gritty. White to milky and granular quartz veining makes up 5% of interval. Minor shearing and related fracturing in dark phyllite.	1-2														
			54.5: S2 @ 70 to c.a. (variable).															
56.2	66.5	QSP Grit	**Quartz-Sericite Phyllite:** Gritty quartz-eye unit, pale grey, medium-grained with strongly foliated to brecciated texture. Strongly foliated and granular texture distinguish unit from blockier felsic volcanic unit. 1-2% pyrite, mainly as medium to coarse cubes.	1-2														
			57.4: S2 @ 55 to c.a.															
			57.4: Minor fault; 2-3cm of pale grey clay gouge, @ 45 to c.a, ~S2.															
			59.3-61.0: Cream to pale green carbonate alteration with modest Cr-mica.															
			62.2: Fault; minor marked by 5cm grey clay gouge.															
			64.7-65.2: Cream to pale green carbonate alteration with modest Cr-mica.															
			64.9: S2 @ 75 to c.a.															
			65.5-66.5: Finer grained, grey and more siliceous.															

Diam[illegible]d Drill Log

DDH No. FC-02-03

Date May 21, [illegible]

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
66.5	68.0	QCSP	**Quartz-Chlorite-Sericite Schist:** Dark grey, moderately siliceous, locally finely layered, gradational with overlying grey grit unit. Weakly "spotted" with sericite overprint. Variable dark chlorite. Upper contact marked by white to grey quartz veining. Sharp lower contact with fault zone.	2														
			68.0: Lower contact, S2 @ 75 to c.a.															
68.0	70.5	Flt	**Fault:** Highly fractured and gougy zone, less intense in places.	2														
			68.0-68.2: Gougy with fine chips.															
			68.2-68.4: Fractured grey siliceous phyllite - gritty ad "spotted".															
			68.4-69.8: Strong carbonate alteration with spotty Cr-mica.															
			68.28-69.80: Recovery ~75%.															
			69.8-70.5: Gouge and gouge breccia.															
70.5	77.7	QCSP	**Quartz-Chlorite-Sericite Phyllite:** Mainly dark grey, finely foliated with minor warping of foliation evident. Grades to argillite.	1-2														
			70.5-70.9: Mainly soft, weakly carbonate altered grit.															
			70.9-71.1: Grey clay gouge, 5cm; quartz vein and minor shear.															
			72.3-72.5: Pale grey, granular to gritty-looking with pale green sericite (?) spots, ~2mm diameter beginning and continuing below. Medium to coarse cubic pyrite continue, only ~1%. S2 @ 40-70 to c.a.															
			73.2: Gougy shear @ 30 to c.a.; S2 @ 60 to c.a.															
			74.4: 2-4cm shear zone/breccia @ 70 to c.a.															
			74.8-76.7: Sharp increase in 1-2cm thick white quartz veins, form a weak stockwork constituting 10% of interval.															
			76.7-76.8: Fault; minor with 7cm of grey clay gouge.															
			76.8-77.7: Well-fractured, mainly QCSP, minor grit. Recovery 50-60%.															
77.7	79.2	QSP	Cream to pale greenish, strongly carbonate-altered, quartz-eyes present - may be grit or felsic unit. Flecks of Cr-mica speckled throughout.	2														
			78.9-79.0: Dark grey QCSP.															
79.2	79.7	QCSP	**Quartz-Chlorite-Sericite Phyllite**. Increased fracturing as fault is approached.															
79.7	80.9	Flt	Strongly gougy **fault**; QBP to 80.3, carbonate-altered QSP to bottom.															
			80.8: Gougy planar slip @ 55 to c.a.															
80.9	89.2	QSP	**Quartz-Sericite Phyllite:** Similar to 77.7-79.2m. Very competent and	2														
		Grit	weakly fractured.											82.0	83.5	8259	102	96
			80.9-81.7: Strong sericite spots, weak reaction to HCl. 81.5: S2 @ 55 to c.a.															
			81.7: Quartz vein along contact, 8cm thick, with minor Cr-mica.															

Diam[illegible] Drill Log

DDH No. FC-02-03

Date May 21, [illegible]

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	to	Sample Number		
			81.8-89.2: Strong carbonate alteration with Cr-mica flecks throughout.															
			85.8-85.9: Minor gougy shear, ~5cm, fractured to 86.1m.															
			88.0-89.2: Moderately fractured.															
			88.6: Strong Cr-mica, fine flecks oriented with S2.															
			89.2: Weakly gougy along contact; S2 @ 65 to c.a.															
89.2	90.0	QSP Grit	**Quartz-Sericite Phyllite-Grit:** Pale grey, very similar to above QSP but stronger foliation, strong sericitic spots, weak quartz eyes, and very competent. Continuing weak to moderate carbonate alteration. 1-2% fine-grained and coarse cubic pyrite.	1-2														
90.0	94.6	Arg	**Argillite:** Mainly dark grey to black, fine-grained, silt to mudstone with minor grit; strong to somewhat contorted foliation. Minor sericitic spotting in coarser intervals. Coarse pyrite cubes ~2%. A few minor shear zones scattered. Irregular S2 parallel quartz veins or sweats to 3cm, make up 5-10% of unit.	2-3														
			90.0: Upper contact is sharp, parallel to S2 @ 70 to c.a.															
			90.6: 16cm interval of grey grit, upper contact marked by white quartz vein and 1cm of gouge @ 70 to c.a., lower sharp @ 70 to c.a.															
			91.7-91.9: Fault; strong pale grey clay gouge, lower slip @ 80 to c.a.															
			94.0-94.6: Moderately fractured.															
94.6	96.1	QSP Grit	**Quartz-Sericite Phyllite-Grit:** Pale grey phyllite, some quartz eyes with an indistinct fine to medium-grained texture. Moderate sericite spots and wispy carbonate alteration. Coarse cubic pyrite to 1cm, up to 5%.	3-5														
			95.6: S2 @ 60 to c.a.															
96.1	101.8	Flt	**Fault:** Series of strong clay gouge zones between sheared sediments.	2														
			96.1-98.8: Mainly very soft pale to medium grey gouge, a few cm-size pieces of unsheared grit. Later 8cm thick white quartz vein at 97.2m.															
			98.8-99.3: Grit; a few gougy slips.															
			99.3-99.7: Pale grey clay gouge with mm-size grit clasts.															
			99.7-101.6: Argillite; black, weakly carbonaceous phyllite and gouge. Note very coarse, 1.5cm, pyrite cubes.															
			101.6-101.8: Pale grey clay gouge.															
101.8	103.4	QSP Grit	**Quartz-Sericite Phyllite-Grit:** Pale grey phyllite with fine quartz eyes and the same indistinct granular texture noted above (94.6-96.1m). Sericite spots and generally quite sericitic throughout. Coarse cubic pyrite persists.	3														

Diam[illegible]d Drill Log

DDH No. **FC-02-03**

Date May 21, [illegible]

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Py	% Cp	% Qv	Kf	mt	chl	epi	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
103.4	107.0	Flt	**Fault:** Moderate to strong intensity, locally clay gouge and breccia zone, becoming less gougy (possibly washed away) with poor recovery and numerous pieces of white quartz vein.															
			103.4-104.7: Mainly grey clay gouge with clasts of grit, QCSP, QSP.															
			103.33-104.85: 65% recovery.															
			104.85-105.77: 90% recovery.															
			105.77-106.38: 50% recovery.															
			106.38-106.98: 25% recovery.															
			104.7-107.0: Grey phyllite & white quartz vein rubble with minor sericite-clay gouge.															
107.0	111.7	QSP	**Quartz-Sericite Phyllite:** Pale greenish grey, fine-grained and moderately phyllitic, soft yet competent, with wispy carbonate alteration and scattered flecks of Cr-mica. Also, sericite spots are relatively common.	2														
			108.2: Strong wispy band of Cr-mica in carbonate flooded zone.															
			109.4: S2 @ 60 to c.a.															
			110.2-110.3: Contorted and brecciated.															
			111.0: S2 @ 80 to c.a.															
			111.5-111.7: Clay-altered, very soft shear zone.															
111.7	113.9	Flt	**Fault:** Weak fault zone consisting of narrow gouge zones and soft moderately sheared grit, and quartz biotite-sericite phyllite.	1														
			112.0-112.2: Gougy rubble zone.															
			112.3: Gouge seam, grey clay, 1cm thick @ 75 to c.a.															
			112.3-113.9: Stockwork of thin clay gouge seams.															
113.9	118.9	QSP	**Quartz-Sericite (Chlorite) Phyllite:** Medium grey, locally strongly brecciated and contorted by deformation, generally fine-grained with wispy carbonate alteration throughout.	1-2														
			113.9-114.3: Strongly spotted.															
			115.0-115.9: Moderately fractured.															
			115.9-116.0: Fault: 4cm pale grey clay gouge.															
			116.0-116.2: Strong carbonate alteration.															
			116.7-117.0: White quartz veining, possibly intruded toward end of deformation giving a contorted appearance. This interval is brecciated during deformation and healed. Wispy pale cream carbonate alteration appears to be associated or intensified by quartz veining.															
			117.0-118.87: Strong carbonate altn assocaited with matrix of breccia.															
			END OF HOLE = 118.87 metres															

Diamond Drill Log

DDH No. **FC-02-04**

Company:	**Barker Minerals Ltd**	Date Collared	May 20, 2002
Project:	**Frank Creek**	Date Completed	May 22, 2002
Logger:	Christopher J. Wild. P.Eng.	Northing	54 + 50N
Date:	May 23, 2002	Easting	12 + 20W
Remarks	Test east of Frank Creek showing, area of high chargeability, low resistivity.	Elevation	1140 m

Azimuth	090
Dip	-45
Length	158.19 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
0.0	5.20	**Casing**	Cased through glacial till, no recovery.														
5.20	13.60	**QCSP**	**Quartz-Chlorite-Sericite Phyllite:** Medium to dark grey, variably textured including interbedded darker mud and siltstone, locally tectonically brecciated. Generally spotted texture in fine and medium-grained components. Weak cubic and disseminated pyrite. Local moderate fracturing, otherwise competent.		**1-2**												
			5.2-5.8: Rusted "spots", possibly originally ankerite.														
			6.0-6.8: Finely layered, darker mudstone/phyllite, strong deformation breccia.														
			6.8-7.2: Thicker layered, medium-grey siltstone.														
			7.2-7.6: Fractured around white to grey quartz vein.														
			7.8-8.0: Fracture zone in dark phyllite. 8.7-8.9: Deformation breccia.														
			9.1-9.7: Finely layered dark phyllite.														
			9.7-12.8: Mainly spotted medium grey silty phyllite.										**9.7**	**10.7**	**8260**	**15**	**91**
			12.9-13.1: Increased fracturing in dark phyllite; minor shearing, transitional.														
			13.1-13.6: Siliceous, pale grey deformation breccia, healed.														
13.60	17.10	**QSP**	**Quartz-Sericite Phyllite:** Pale grey, fine to medium-grained, more massive, soft but weakly fractured - possibly felsic volcanic or dyke. A few white porphyroblasts, no quartz eyes noted. Not spotted but **strongly calcareous**; traces of Cr-mica, 2-3% finer-grained pyrite.		**2**								**16.1**	**17.1**	**8261**	**108**	**86**
			13.6: Strong reaction with HCl when powdered. Contact @ 50 to c.a.														
			14.3-14.7: Darker grey, more siliceous, healed breccia - host to calcareous unit.														
			14.7-15.5: Moderately fractured, medium to pale grey, non calcareous, locally spotted with quartz eyes clearly visible in the spots (ankerite altn?).														
			15.5-16.1: Limonitic fractures, end at 16.1 - becomes very massive.														
17.10	43.60	**QCSP**	**Quartz-Chlorite-Sericite Phyllite:** Medium to dark grey, fine to medium-grained, mixed interval interlayered silty and muddy units. Variable rock quality, more fractured in darker phyllite. Weak pyrite, usually cubes. Spotted, likely the broad ankerite altn; no reaction with HCl but rusted out in oxidized top of hole.														
			17.1-21.3: Spotted grey phyllite; weakly fractured, competent; variable breccia texture.														
			21.3-21.9: Fracture/shear zone, minor white quartz vein, rubbly; S2 @ 45 to c.a.														
			21.9-26.0: Generally darker, more fractured and clearly healed tectonic breccia throughout. Interlayered dark and medium grey phyllites.														
			23.0: Cream-coloured, wispy carbonate alteration with flecks of Cr-mica, 5cm @ 75 to c.a. (S2).														

Diam d Drill Log DDH No. FC-02-04 Date May 23,

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
			29.7-29.9: Well-fractured shear zone, 2cm white quartz vein, 2cm grey gouge seam.														
			26.0-27.3: Medium grey, siliceous unit, possibly weakly gritty, competent.														
			27.3: S2 @ 55 to c.a.														
			27.3-29.7: Strongly sheared, low angle to c.a., moderately to strongly fractured, not gougy. Mainly dark phyllite with lesser amounts of silty component.														
			27.3-28.0: Shear zone.														
			29.0-29.7: Shear zone.														
			29.7-30.2: Competent healed tectonic breccia.														
			30.2-31.1: Shear zone.														
			31.7: 10cm section of 10% pyrite cubes.														
			31.1-31.5: Fine undulating layers, pale and dark grey, strongly sericitic; l0 @ 50 to c.a.														
			31.5-33.4: Medium-grey, silty, spotted, weakly fractured and competent.														
			33.4-36.0: Medium-grey, more foliated, locally sericitic with pale spots. Much more sheared, local gouge intervals and tectonic breccia.														
			36.0-37.1: Increased fracturing related to brittle white quartz veins; minor fault.														
			37.6: Minor fault, weakly gougy along slip @ 30 to c.a.														
			37.9-38.2: Cream-coloured, wispy carbonate alteration, no obvious Cr-mica.														
			38.2-39.8: Fine-grained, finely layered, contorted and brecciated, quite soft.														
			39.8-40.8: Cream-coloured, wispy carbonate alteration, minor Cr-mica.														
			40.8-43.6: Well-foliated, partially carbonate altered phyllite to medium grey silty phyllite with coarse deformation breccia and some dark phyllite layers. Increased pyrite cubes.		3-4												
43.6	46.0	QSP	**Quartz-Sericite Phyllite:** Pale cream to greenish-grey, fine-grained phyllite with wispy carbonate alteration and increased bright green Cr-mica throughout. Soft but competent.		<1								43.6	44.6	8262	57	72
			45.2: Vuggy white quartz vein, 3.5cm thick, @ 20 to c.a.														
46.0	47.4	Arg	**Argillite:** Dark grey to black phyllite, moderately carbonaceous, contorted layering and breccia; moderately fractured.		2												
			46.4: 5-10cm shear zone, polished slips, well-fractured with large pyrite cubes.														
			46.4: S2 @ 55 to c.a.														
47.4	50.5	QSP	**Quartz-Sericite Phyllite:** Pale cream to greenish grey phyllite, siltstone with moderate spots - ankerite and carbonate alteration, Ranges from possibly gritty near the top to more blocky breccia lower. Fine-grained, silty appearance.		1-2												
			47.4-48.4: Fractured along c.a.														
			49.1: Fault; minor with 1-2cm of grey clay gouge.														
			49.4: Fault; minor with 4cm of greenish grey clay gouge @ 55 to c.a.														
			49.3-50.5: Coarse deformation breccia with fine-grained cb-altered clasts.														

Diam[illegible]d Drill Log

DDH No. **FC-02-04** **Date** May 23, [illegible]

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
50.5	59.0	Ls Bx	**Limestone Breccia:** Pale to medium grey limestone clasts in darker grey matrix, largely clast-supported; clasts elongate along foliation, ranging from 1mm to >4cm, typically 1-3cm. Matrix and some siliceous clasts are not calcareous.A few clasts of strongly carbonate-altered phyllite noted, esp @ 52.8m. Interval is weakly fractured and competent. Pyrite continues to be cubic and weak.		1-2												
			53.8-54.4: Pale greenish phyllite, wispy, weakly calcareous alteration.														
			56.4: S2 @ 55 to c.a. 57.0-57.6: Dark muddy matrix, fractured.														
			57.9-58.1: Tan carb altn, more phyllitic section.														
59.0	64.1	QSP-Bx	**Quartz-Sericite Phyllite-Breccia:** Similar to above but medium grey non-calcareous and more siliceous (harder) phyllite clasts in deformation breccia. Clast sizes, sorting and angularity are similar to above; ie, poorly sorted and subrounded, less well-defined, less angular and slightly smaller.		1-2								60.0	61.0	8263	10	43
64.1	64.9	Flt	Dark grey, very soft clay and sericite gouge. **Fault** appears to be conformable.														
64.9	65.8	QSP-Bx	**Quartz-Sericite Phyllite-Breccia:** As before (59.0-64.1m), more rubbly.														
			65.4-65.8: Well-fractured, strongly sheared.														
65.8	67.9	Arg	**Argillite:** Dark grey, fine-grained, tectonically brecciated and healed. Dark and contorted, competent except at top of interval.		2												
67.9	69.3	Arg-Bx	**Argillite-Breccia:** Strong tectonic breccia, as described above. Darker, fine-grained argillite matrix. Weakly fractured and very competent. Increasing coarse cubic pyrite. 69.0: S2 @ 50 to c.a., curved.		3-4												
			69.1-69.3: Contorted argillite.														
69.3	70.0	QSP	**Quartz-Sericite Phyllite:** Medium grey, fine to medium-grained, siliceous and hard, possibly gritty or weakly deformed dyke. Brecciated and healed at lower contact. Pyrite cubes continue. Strong ankerite spots. Competent.		3												
70.0	71.2	Arg-Bx	**Argillite-Breccia:** Dark grey to black, fine-grained, coarse clasts in deformation breccia. Soft yet competent. Coarse cubic pyrite continues.		3												
71.2	71.9	Flt	**Fault:** Strongly gougy, black, moderately carbonaceous argillite. Approx 80% recovery, very mushy gouge at lower contact. Undulatory upper slip @ 30 to c.a.														
71.9	73.3	Arg-Bx	**Argillite-Breccia:** Deformation breccia, as above.		2-3												
73.3	76.7	Ls Bx	**Limestone Breccia:** Well-layered, with elongated pale grey limestone clasts in darker, less calcareous matrix. Same as limy interval (50.5-59.0m). 2cm white quartz veins common. More layered, less bx'd.		2												
			75.9: S2 & I0 @ 60 to c.a.														
			76.7: S2 & I0 @ 25 to c.a., curved.														

Diam[illegible]d Drill Log DDH No. **FC-02-04** Date May 23, [illegible]

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number		
									Alteration score 1 - 5								
76.7	86.5	QCSP	**Quartz-Chlorite-Sericite Phyllite:** Mixed interval, dark and medium grey, interlayered siltstone and mudstone phyllites, generally deformed and brecciated. Locally finely layered. Generally weakly fractured and quite competent. Pyrite continues somewhat weak. Increased tan carb with quartz veining.		1-2												
			76.8-77.1: Several S2-parallel white quartz veins, irregular, ~1cm thick.														
			77.2-77.6: Mainly pale grey gritty-looking phyllite, well-laminated with quartz eyes.														
			77.6-78.1: Strong deformation breccia, as described previously.														
			78.1-79.8: Medium grey, fine quartz-eye grit with layers of mudstone that have flowed around more rigid grit during deformation.														
			79.8-80.2: White barren quartz vein; moderately fractured.														
			80.2-80.8: Grey quartz-eye grit.														
			83.6-84.2: Minor shear and fracture zone, rubbly with white 20% white quartz.														
			84.2-86.5: Darker, less gritty phyllite, minor breccia and quartz veining.														
			84.8: S2 @ 60 to c.a.														
86.5	86.9	Flt	**Fault:** Intense clay gouge, very sharp slips, lower @ 70 to c.a. (S2).														
86.9	103.2	Arg	**Argillite:** Mixed interval (same as 76.7-86.5m), dark and medium grey, interlayered siltstone and mudstone, variable deformed and brecciated. Locally finely layered. Generally weakly fractured and quite competent. Pyrite continues somewhat weak. Patchy tan carb in qtz vns.		1-2												
			89.3-91.7: Similar to above but more clearly breccia with lapilli shaped silty clasts in muddy matrix.														
			91.7-93.2: Well-layered and less intense breccia textures.														
			93.1: S2 @ 75 to c.a.; highly variable from 40-80 to c.a.														
			93.1-93.7: Pale green, strongly sericitic dyke (?), upper contact @ 60 to c.a., sheared, lower contact appears to fade into darker phyllite.														
			93.7-97.3: Finely layered, soft silty and muddy interval; S2 @ 50 to c.a. (95.4m)														
			97.3-98.1: Sharp increase in frequency of cm-size S2 parallel quartz veins, associated with carbonate alteration and Cr-mica.														
			98.1-103.2: Mixed interval with quartz veining, well-foliated grits and dark phyllite.														
			99.2-99.5: Medium grey, fine-grained, hard with 5-10% f-gr pyrite. Both contacts sharp @ 60-65 to c.a., parallel to S2.		5-10												
103.2	107.5	Arg	**Argillite:** Dark grey, finely layered dark phyllite, similar to some finely layered section in overlying interval. 5% white quartz veining, frequently S2-parallel. Not strongly pyritic, moderately fractured and less competent.		2												

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
			Sharp increase in quartz veining to 20% of interval. Host rock is medium grey phyllite, locally finely-layered, moderately fractured.														
			106.8-107.5: Little quartz veining, finely layered phyllite, increased fracturing.														
107.5	108.7	Flt	**Fault:** Mixture of QCSP and fine white quartz rubble, and grey clay gouge.														
			106.98-108.20: 65% recovery.														
			108.7: Rough lower slip @ 30 to c.a.														
108.7	114.3	QCSP	**Quartz-Chlorite-Sericite Phyllite:** Medium-grey phyllite, locally well-layered (no significant breccia textures), intruded by increasing white quartz veining.		3-5												
			109.1-109.9: Weak wispy tan carbonate-alteration, no obvious associated Cr-mica; significant silicification. Somewhat more pyrite.														
			109.9-111.1: Well-layered, 5% pyrite.														
			111.1-111.5: White quartz veining, approx 50%. Minor tan carbonate in quartz.														
			112.3-112.8: Mainly white quartz vein, irregular contacts.										112.3	112.8	8264	33	371
			112.8-114.3: Moderate to strong fracturing; dark grey phyllite and quartz.														
114.3	127.8	Flt	**Fault:** Black, moderate to strong graphite, minor quartz in gouge. Variable recovery.														
			113.69-114.60: 55% recovery.														
			115.21-115.82: 90% recovery.														
			114.3-115.3: Finely layered dark phyllite, mainly rubble.														
			115.3-116.0: White quartz rubble in dark graphitic clay gouge.														
			116.0-116.5: Highly sheared, contorted and locally graphitic and gougy dark phyllite.														
			115.82-117.35: 65 % recovery.														
			117.35-118.26: 75% recovery; mainly graphitic gouge and white quartz rubble.														
			118.26-119.18: 87% recovery; more graphitic gouge, less quartz rubble.														
			119.18-120.09: 45% recovery; graphitic rubble and gouge.														
			120.09-121.92: 60% recovery; graphitic gouge.														
			121.92-123.44: 20% recovery; graphitic gouge.														
			123.44-124.66: 80% recovery; mainly graphitic gouge, more solid graphitic phyllite.														
			124.66-125.27: 65% recovery; minor quartz rubble, graphitic phyllite and gouge.														
			125.27-125.88: 100% recovery; good graphitic phyllite to 125.6m, then gouge.														
			125.88-126.80: 100% recovery; dark phyllite blocks in gouge.														
			126.80-128.02: 80% recovery; slight increase in gouge.														
127.8	130.2	QSP	**Quartz-Sericite Phyllite:** Pale greenish grey, highly sheared, locally gougy phyllite. Strong carbonate & Cr-mica alteration, locally excellent Cr-mica.														
			127.8-128.2: Coarse flakes of bright green Cr-mica.														

Diam(d Drill Log   DDH No. FC-02-04   Date May 23,(.

Alteration score 1 - 5

Depth (m) From	Depth (m) To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	Interval (m) to	Sample Number	Cu ppm	Zn ppm
			128.2-128.4: Mottled white and tan quartz-carbonate vein, with grey phyllitic parting and strong Cr-mica along parting and contacts.														
			128.7-129.1: Strong Cr-mica; minor quartz vein, 2cm thick.														
			129.6-130.2: Fault; sheared and fractured with 12cm gouge @ 129.8m.														
130.2	144.6	QCSP	**Quartz-Chlorite-Sericite Phyllite:** Mixed interval, medium to dark grey, well-layered, heterogenous phyllite; locally dark biotite-rich or carbonate altered and sericitic. Weak pyrite, weakly to moderately fractured.		1												
			130.2-131.3: Spotty tan carbonate alteration.														
			132.2: 4cm gouge seam @ 40 to c.a.														
			133.8-134.1: Strong deformation breccia textures.														
			134.1-134.3: Strong "spotted" texture; several replcaed with pyrite.														
			134.6-136.0: Significant white quartz veins (5-10%), mottled with carbonate?														
			137.2-137.4: White quartz and mottled carbonate vein 12cm thick, @ 30 to c.a.														
			138.3: Strongly deformed unit with variable S2 @ 60 to c.a.														
			139.0: 5cm band of strong tan carbonate alteration, @ 70 to c.a. (S2)														
			139.2: 4cm gougy series of slips @ 70 to c.a.														
			139.8: 16cm band of strong tan carbonate alteration @ 75 to c.a.														
			141.6-142.2: Strong deformation breccia textures.														
144.6	147.7	QSP	**Quartz-Sericite Phyllite:** Tan to pale grey, fine grained, strongly carbonate-altered, strongly sericitic phyllite. Scattered quartz eyes, <1mm in size; traces of wispy Cr-mica. Unit exhibits the same deformation textures as the surrounding dark phyllites. Competent and weakly fractured. Pyrite?										144.6	146.1	8265	15	71
			144.6: Sharp but irregular upper contact @ 60 to c.a.														
			145.2: S2 @ 80 to c.a.														
			146.2-143.6: 50% recovery, thin chips and rubble.														
147.7	158.2	QCSP	**Quartz-Chlorite-Sericite Phyllite:** Strong deformation breccia in quartz-biotite phyllite. Generally tight, clast-supported with section of fine and coarse clasts. Most clasts are pale grey in dark muddy matrix, giving the core a striking appearance. Rootless isoclinal folds are locally apparent. Core is competent and weakly fractured. Continuing weak pyrite. Weakly calcareous.		<1												
			154.3-154.6: Pale grey quartz-eye grit unit, sharp contacts parallel to S2 @ 80 to c.a.														
			157.2: Tight fold, likely F2 due to lack of folding of S2.														
			END OF HOLE = 158.19 metres														

Diamond Drill Log

DDH No. FC-02-05

Company:	**Barker Minerals Ltd**	Date Collared	May 22, 2002
Project:	**Frank Creek**	Date Completed	May 24, 2002
Logger:	Christopher J.Wild. P.Eng.	Northing	55 + 60N
Date:	May 25, 2002	Easting	15 + 00W
Remarks	From 2nd switchback, D Road, toward "Stringer Outcrop".	Elevation	1115 m

Azimuth	075
Dip	-45
Length	158.50 m
Core Size	NQ

Tests

Depth	Dip
n/a	

altn 1- weak, 5 strong

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
0.0	4.57	Casing	Cased through glacial till, no recovery.														
4.57	5.10	Ovbn	Rubble, rounded, likely not from bedrock.														
5.10	8.90	Arg	**Argillite:** Dark grey, fine-grained, finely layered, rusty, dark phyllite, originally mudstone. Coarse blebby and cubic pyrite, some occurring along distinct layers. *Dark limonite along fractures, in pyrite, and in 1mm spots.* Relatively good rock quality, especially for top of hole. No significant deformation breccia noted.		4	<1	tr										
			5.4: 1cm band of cubic to blebby pyrite @ 60 to c.a. (S2).														
			5.5: Strong crenulation cleavage (S3) @ 30 to c.a.														
			6.3: Wispy chalcopyrite-sphalerite patch, 0.5 x 2cm, separate from pyrite.										**6.00**	**6.30**	**1LF05-12**	**1.14%**	**1050**
			8.3: Wispy chalcopyrite in strong dark limonite band, irregular.										**6.60**	**6.62**	**1LF05-14**	**7130**	**4290**
8.90	14.50	QSP	**Quartz-Sericite Phyllite:** Sheared upper contact, 5cm thick @ 80 to c.a., into much paler grey, strongly sericitic, finely layered fine-grained phyllite. Pyrite continues along S2, mainly as medium cubes. Moderately fractured, again little deformation breccia.		3-5	tr							**9.15**	**9.45**	**1LF05-13**	**829**	**988**
			9.7, 10.0: 5-10mm bands of 25-50% cubic pyrite, @ 80 to c.a.														
			10.5-10.9: Weak healed breccias, likely deformational.														
			10.9: End of significant surface oxidation.														
			11.4: Irregular 1-4cm band of mainly cubic pyrite.														
			13.1-13.5: Cracked and rehealed glassy, medium grey quartz. Minor sericite, pyrite and carbonate form very weak stockwork in quartz.														
			13.6-13.9: Dark grey, well-layered Arg, as above; lower contact @ 55 to c.a. (S2).														
			13.9-14.5: Pale grey phyllite, largely quartz-flooded.														
14.50	21.00	Arg	**Argillite:** Dark grey with fine pale grey interlayers, strongly contorted and quite pyritic, much as before (5.1-8.9m). Pyrite most often occurs as cubes and blebs in pale grey layers.		5	tr											
			16.5: S2 @ 75 to c.a														
			19.2: S2 @ 0 to c.a.; F3 closure with associated tight parasitic folds.														
			19.5-19.6: >25% pyrite, mainly as disassociated fine cubes, along layer.														
			19.8: 5 cm band of cubic pyrite.														

Diam[illegible]d Drill Log   DDH No. FC-02-05   Date May 25,[illegible]

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
			20.2: **4-5cm band of semi-massive sulphides**, 25% pyrite, 3-5% cp, 1% sphalerite, trace galena, all intimately related, not magnetic. Band @ 60 to c.a.		25	4	1	tr					19.5	21.0	8266	1970	2268
													19.50	21.00	2LF05-16	1885	2850
			20.35: 5cm series of 3 pyrite-chalcopyrite-sphalerite bands.		15	2	0.5	tr					20.20	20.30	3LF05-15	8950	4080
21.0	32.9	QSP	**Quartz-Sericite Phyllite:** Pale grey, weak to strong layering, locally well-mineralized with pyrite and minor chalcopyrite. Well-layered sections grade into darker phyllite. Interval is weakly fractured and competent.		5	tr	tr						21.0	22.8	8267	2751	1252
													22.80	23.70	1LF05-23	2560	1125
													23.70	24.70	1LF05-24	705	1220
			21.2-22.8: Coarse disseminated and wispy pyrite, chalcopyrite with minor to trace sphalerite and galena. **Strong cp @ 21.2-21.4, 22.5-22.6m.** Associated with grey quartz and QSP, possible feeder type mineralization.										24.70	25.70	1LF05-25	577	1250
													25.70	26.50	1LF05-26	878	512
			23.1-23.6: Fault: 10-20% recovery, 3cm gouge, rubbly.														
			24.6: 1-2cm clay gouge and pebble breccia fault @ 65 to c.a.														
			24.8: Minor chalcopyrite and trace sphalerite, weak minl continues.														
			26.1: White quartz vein with tan carbonate selvages, 1-2cm thick @ 50 to c.a.														
			26.6: 7cm band of 20% cubic pyrite @ 80 to c.a.										26.5	28.0	8268	807	872
			26.8: Cubic pyrite and **5% blebby cp** associated with grey fist-sized quartz knot.										28.0	29.5	8269	908	1349
			27.0-27.1: 1cm layers of dark phyllite with intervening pale grey phyllite with cubic pyrite and minor blebby cp.										29.5	31.0	8270	4935	1255
													31.0	32.9	8271	539	502
			27.9: Pyrite band with minor blebby cp and fine reddish-brown sphalerite.														
			28.3: Blebby pyrite with minor wispy cp and fine reddish-brown sphalerite.										29.50	30.40	3LF05-33	7220	2380
			28.5: 1-2cm band of coarse cubic pyrite and fine red-brown sphalerite @ 45 to c.a.										30.40	31.00	2LF05-34	815	254
			29.4-30.4: Broad zone of mineralization, disseminated and in bands, made of of mainly coarse pyrite with minor wispy cp, and traces of reddish-brown sphalerite.		10	1	tr										
			29.6: Band of chalcopyrite, trace sphalerite, 0.5cm thick @ 60 to c.a.														
			30.2-30.3: Wispy cp associated with coarse cubic pyrite, S2 @ 45 to c.a.														
			31.7: 1cm band of pyrite, lesser cp and minor sphalerite.														
			32.3: Wispy band 1-2mm thick of cp and minor sphalerite.														
			32.9: Lower contact with thinly layered unit @ 55 to c.a.														
32.9	37.2	Arg	**Argillite:** Dark grey with fine pale grey interlayers, strongly contorted and quite pyritic, much as before (5.1-8.9m). Foliation (S2) is contorted, folded (F3) with developing crenulation cleavage (S3). Pyrite most often occurs as cubes and blebs in pale grey layers.														
			37.4: Crenulation cleavage (S3) @ 25 to c.a.; variable.														
			33.3: 3-4mm white quartz vein, mottled with tan carbonate, @ 65 to c.a.														
			34.2: Clearly outlined F3 antiform verging uphole to the west.														
			35.5: 1cm band of coarse pyrite, minor reddish-brown sphalerite.										35.0	36.0	8272	55	768
			36.1: F3 crenulation fold, cleavage @ 30 to c.a., S2 @ 35-40 to c.a., opposite.														
			36.9-37.2: Weakly gougy shear zone; bottom slip @ 30 to c.a.														

Diam d Drill Log DDH No. FC-02-05 Date May 25, .

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
37.2	50.3	QSP	**Quartz-Sericite Phyllite:** Pale grey, fine to med-grained, locally quartz-eye and ser-rich. Ser may represent altn of felsic volcanic, or regional metamorphism of pelitic rocks. Locally mineralized, mainly pyrite, lesser cp, minor sph and traces of galena. Interval is relatively competent, fracture zones noted.		2-3												
			37.8-37.9: Pale grey swelled clay gouge with QSP clasts.														
			38.8-38.9, 39.1-39.2: **Strongly mineralized section**, approx 10 cm thick, consisting of blebby to cubic pyrite, patchy to finely disseminated chalcopyrite, and minor disseminated sphalerite in a soft sericite-chlorite host with occasional elongate clasts 1-2cm in length.	tr	15	3-4	1	tr	2	1	4	1	38.8	39.2	8273	2104	1671
													38.70	38.90	3LF05-43	3970	2580
													38.90	39.20	3LF05-42	1725	748
			39.0-39.5: Weakly gougy and sheared.														
			40.0: S2 flattens from 50 to 20 back to 50 to c.a. on an open monocline.														
			40.3: Minor gougy shear.														
			40.7-42.0: Well-layered, 1-2mm, pale grey throughout, with broad open folding and developing axial planar cleavage (S3).														
			41.6: S3 @ 30-50 to c.a.; verging to antiform to west.														
			44.6, 44.8: Gougy shears, 3-5cm thick, generally parallel to S2.														
			47.0-47.1: White mottled quartz-carbonate blowout with minor pyrite and sphalerite.														
			48.7: Weakly mineralized zone (6-7cm)consisting one narrow band of pyrite-minor sphalerite and 1-2cm band of mainly pyrite, minor sphalerite, cp, and galena. Disseminated mineralization between bands.										48.7	50.3	8274	342	750
			49.7: 10cm weakly gougy fracture zone; S2 @ 75 to c.a.														
			50.1: Patchy pyrite, minor assoc sph, in and adjacent to grey quartz vein.														
50.3	55.6	DS	**Disseminated Sulphide Zone:** Significant wispy, disseminated to semi-massive sulphides in a siliceous felsic unit, possibly volcanic breccia. Sulphides include pyrite, mainly fine-grained cubes, as wisps and patches, minor scattered patchy to disseminated cp, less reddish sphalerite and traces of bluish silver galena. Concentrations of each are highly variable, with py-cp and py-sph affinities. Often pyrite occurs without cp or sph.		10	<1	<<1	tr					50.3	51.8	8275	1173	501
													51.8	53.3	8276	1514	1380
													53.3	54.8	8277	2315	1926
													54.8	55.6	8278	3612	4145
			50.3: S2 @ 80 to c.a.														
			50.3-50.9: 15% sulphides, mainly pyrite, wispy discontinuous layers.		15	tr	tr										
			51.3-51.5: 10-15% pyrite.		15	tr	tr										
			51.9-52.3: Wispy py-cp, fine-grained sph.		12	1	<1										
			51.9: S2 @ 60 to c.a.														
			52.4-53.5: Wispy py-cp, minor sph.		10	1	<1										
			52.5: S2 @ 45 to c.a., continues to flatten.														
			53.2: S2 @ 30 to c.a.														
			53.5-54.2: 1-2% disseminated pyrite.														
			52.4: S2 @ 20 to c.a.														
			52.4-55.6: Increase in sulphides, esp @ 55.0-55.3m, including cp & sph, locally semi-massive. S2 @ 30 to c.a., at 55.2m		20	2	1	tr									

Diam d Drill Log DDH No. **FC-02-05** Date May 25,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
55.6	56.3	SMS	**Semi-Massive Sulphide Zone:** close to 50% sulphides, still pyrite-dominant, fine-grained, with patchy to wispy cp in py and fine reddish sph with py and sometimes cp, fine galena more isolated. Zone is not magnetic.Host is very siliceous, not calcareous, much is grey quartz. Difficult to determine orientation of layering or S2, quite contorted and folded.		25	3	2	<<1					55.6	56.3	8279	3189	3522
													55.6	55.8	2LF05-38	666	2500
													55.8	56.3	3LF05-39	4650	4000
56.3	57.9	DS	**Disseminated Sulphide Zone:** Significant wispy, disseminated to semi-massive sulphides in a siliceous felsic unit, more granular than before. Far weaker than hangingwall section. Sulphides include pyrite, mainly fine-grained cubes, as wisps and patches, minor scattered patchy to disseminated cp, less reddish sphalerite and traces of bluish silver galena. Concentrations of each are highly variable, with py-cp and py-sph affinities. Often pyrite occurs without cp or sph. Later quartz veins more common, less structurally contorted with abruptly steeper foliation.		10	1	0.5	tr					56.3	57.9	8280	861	821
			56.3-56.5: Well-gractured, more white quartz, not mineralized, 1cm band of py-sph.														
			57.4: Good S2 @ 75 to c.a.														
57.9	68.5	QSP	**Quartz-Sericite Phyllite:** Pale grey, somewhat more granular-looking with fine quartz eyes, more massive and little of the good felsic banding seen in the hw. Irregular quartz veining continues with less sulphides, mainly medium-grained cubic pyrite.										57.9	59.4	8281	94	233
			58.2: S2 @ 75 to c.a														
			59.4: White quartz vein with mottled tan carbonate selvages, 2cm thick, @ 50 to c.a., crosscutting S2.														
			61.0-61.1: Irregular stockwork of coarse cubic pyrite, up to 1cm thick.										60.70	61.10	1LF05-30	199	176
			61.1-68.5: Moderately fractured section with considerable irregular grey quartz veining, likely as a stockwork. Folded, contorted and locally sheared - looks patchy on several centimetre scale. Host unit is strongly sericitic.														
			63.6-63.8: Weakly gougy shear zone @ 50-55 to c.a., minor.														
			64.2: 2cm patch of pyrite with minor associated cp + sph.														
			64.2-64.4: Brittle quartz vein fracture zone.														
			65.0: 1.0 x 0.5cm patch of cp.														
			65.3-65.5: F3 with S2 flattening to 0 to c.a., broad and open, no vergence sense.														
			65.6: Minor 1cm grey clay gouge @ 70 to c.a.														
			66.1-68.5: Sharp increase in coarse pyrite cubes.														
68.5	82.1	QSP	**Quartz-Sericite Phyllite:** Much more uniform section, massive and granular-looking with fine glassy quartz eyes in pale greenish sericitic matrix, weakly fractured and not contorted. Planar bands of phyllite make up less than 10% of interval. Looks sedimentary or volcaniclastic. Increasing Fe-carb altn.										68.50	68.90	1LF05-27	295	386
													69.60	69.75	1LF05-17	381	354
			71.2: S2 @ 63 to c.a., strongly sericitic plane.														
			72.2: Band of patchy and disseminated pyrite - cp, both occur in phyllite along S2 and in irregular crosscutting grey quartz veinlet. S2 @ 45 to c.a.														

Diam d Drill Log DDH No. **FC-02-05** Date May 25,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
			72.3: Weak patch of chalcopyrite-pyrrhotite, 2 x 10mm, weakly magnetic.										73.20	73.55	1LF05-18	582	232
			73.6-74.2: Blotchy healed bluish grey quartz in swirled and deformed phyllite, only minor pyrite associated with quartz.										74.60	75.60	1LF05-19	122	94
													75.00	75.08	3LF05-19A	364	484
			75.2: Band of wispy po-py-cp mineralization, 2-3cm thick, @ 25 to c.a.														
			75.4-75.5: Botchy quartz veining, as above, with minor pyrite.														
			75.7: Band of semi-massive sulphide, much of it washed out leaving limonitic vugs; 5% patchy and wispy cp, 10-15% weakly magnetic po+py,														
			77.4: Grains of pyrite, some cubes, with very minor sph in band @ 60 to c.a.														
			79.0: Three bands of fine to mediumgrained py+po, weakly magnetic, @ 70 to c.a.										77.8	79.3	8282	130	124
			79.6: Band of py-po-sph-cp, 0.5cm thick, @ 75 to c.a.														
			79.7-79.9: Zone of granular (cubic) pyrite stockwork with minor sph.														
			80.5: 1cm band of medium-grained granular pyrite, traces of reddish sph.														
82.1	87.6	QSP	**Quartz-Sericite Phyllite:** Pale, slightly greenish breccia, locally blocky becoming finer in the centre.														
			ranges up to 5cm, becoming finer downsection with introduction of some vein quartz. Weak sulphides.														
			84.0-86.3: Uniform 1cm felsic breccia clasts, clast-supported in felsic matrix. Sericite common with locally abundant bright green Cr-mica, not obviously carbonate-altered. Only minor pyrite.														
			84.4-84.7: Strong patches of micaceous Cr-mica.														
			86.3-87.6: Similar to 82.1-84.0m section.										85.50	85.70	1LF05-20	47	82
87.6	87.7	Flt	Minor gougy **fault**, poker-chip phyllite, @ 80 to c.a.														
87.7	90.2	QSP	**Quartz-Sericite Phyllite:** Medium-grey, locally "spotted" with sericite or ankerite, well-layered gritty section and finer silty units. Looks sedimentary.		2												
90.2	97.8	QSP	**Quartz-Sericite Phyllite:** Heterogeneous section of pale grey phyllite, with fine quartz-eye grits, pale greenish sericitic intervals, locally brecciated with dark grey quartz vein material. Generally quite compentent and weakly fractured. Breccia appears restricted to quartz vein episode, much later than early deformation breccias.		2								90.2	91.2	8283	48	256
			92.3-93.2: Bluish-grey quartz vein and grey phyllite breccia.														
			95.1-95.7: Vein breccia, as above.														
			95.2: Wispy patch of Cr-mica, associated with patchy pyrite. S2 @ 70 to c.a.														
97.8	136.5	QSP-Bx	**Quartz-Sericite Phyllite:** Pale and medium grey, gritty to fine-grained with layers (beds) greatly attenuated with less plastic gritty units forming clasts aligned with S2 in finer more plastic matrix. Textures vary depending on the proportion of fine and coarse units.		2												

Diam d Drill Log DDH No. FC-02-05 Date May 25,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
			97.8-102.3: Pale grey, granular-looking gritty clast deformation breccia.														
			96.1: White quartz vein, minor tan carbonate, 7cm thick @ 30-40 to c.a.														
			98.4: Blebby pyrite surrounding a grey quartz vein piece.														
			100.2: S2 @ 60 to c.a.														
			100.9-101.2: Mod to strongly fractured, good crenulation lineation on S2.														
			102.3-105.9: Much more finely layered, contorted and aligned clasts, darker grey reflecting fine-grained nature.														
			103.6: Grey quartz veins are tightly folded within chaotically folded phyllite.														
			105.2: S2 @ 80 to c.a.														
			105.9-109.7: Pale grey, coarse clasts, mostly fine grit. Fine grit is medium grained with good quartz eyes <1mm in diameter in sericitic groundmass. Weak greenish sericite, weakly spotted, possibly weak carbonate alteration. Weak cubic pyrite.		1-2												
			109.7-113.0: Grey fine quartz eye grit, weakly brecciated, esp in central part.														
			112.4-113.2: Good tan carbonate alteration, becomes spotted.														
			113.0-136.5: Heterogeneous, less gritty strong deformation breccia. Continuing very competent, weakly fractured, 100% recovery.														
			113.0: 1cm gougy shear @ 45 to c.a.														
			118.5: 5cm grey clay gouge, @ 70 to c.a.														
			119.9-120.2: Pale green dyke, relict felspar phenocrysts. Strong carbonate alteration with associated Cr-mica most noticeable near the contacts. Upper contact @ 60, lower @ 45 to c.a. Weakly fractured but soft, sericite-calcite.														
			121.4: 10cm grey clay gouge fault, @ 40 to c.a.														
			122.4-123.3: Strong breccia texture with pale clasts, poorly sorted, subrounded and matrix-supported.														
			128.2-128.8: Pale grey, strong irregular quartz eyes, clearly conglomeratic (grit-size, max 3mm dia.), fining down (graded bed?).										124.0	125.0	8284	30	152
			128.8-129.2: Highly deformed with introdiction of blue-grey vein quartz.														
			129.3-129.5: Pebby blue-grey quartz veining crosscutting S2 @ 10 to c.a.														
			130.0-130.1: Shear zone, two gougy seams in broken rock, @ 70 to c.a.														
			130.5: S2 @ 60 to c.a.														
			130.5-136.5: Gradually becoming more gritty.														
136.5	139.9	QSP	**Quartz-Sericite Phyllite:** Grey phyllite, strongly sericitic, with sharp increase in number and degree of gougy shear zone and faults. More sheared grit, less true phyllite, weak quartz eyes. Far less competent, moderate to strong fracturing, locally recovery <90%.														
			136.8-137.0: Strong shear zone, 6cm gouge @ 50 to c.a., undulatory.														
			137.4-138.7: Hard and competent, partially silicified, some quartz veining.														

Diamond Drill Log DDH No. **FC-02-05** Date May 25,

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
139.9	151.9	Flt	**Fault:** Broad zone of strong grey clay gouge, gouge breccia, strong brittle shearing superimposed on mainly deformation breccia. Strongly sericitic, local carbonate alteration. Few sections of competent rock.		1												
			139.90-141.43: 55-60% recovery.														
			140.1-141.6: Mainly gouge, intense part of fault.														
			141.6-143.3: Well-fractured, sheared with minor gouge.														
			141.43-142.34: 85% recovery.														
			145.6-146.8: Several strong gouge sections, strong shear zones.														
			143.26-144.48: 55-60% recovery.														
			144.48-145.69: 50-55% recovery														
			145.69-146.91: 90% recovery.														
			146.91-147.83: 90% recovery.														
			147.8-150.4: Strongly sheared, 50% gougy, soft, can cut or crumbled with knife.														
			150.4-151.9: Strongly sheared, crumbly but not strongly gougy.														
			151.2: S2 @ 70 to c.a., appears to be parallel to fault zone.														
151.9	158.5	QSP Grit	**Quartz-Sericite Phyllite:** Pale greenish grey, carbonate-altered with scattered flecks of Cr-mica, in a fine to medium-grained quartz-eye grit. Mostly weakly fractured, but several shears and fracture zones noted. Weak cubic pyrite.		1								153.0	154.0	8285	42	93
			151.9-152.1: Strong Cr-mica. Upper shear @ 35 to c.a.														
			152.1-152.7: Rubbly fracture zone, bottom 20cm mainly white quartz vein.														
			155.2: Mushy gouge contact between QSP and dark mudstone phyllite, over 6cm, @ 70 to c.a.														
			155.2-155.3: 10cm slice of dark grey to black, well layered mudstone @ 70 to c.a.														
			156.0-156.4: Interfingering dark mudstone in carbonate-altered QSP. Contacts are sharp and @ 85-90 to c.a.														
			156.8-157.7: Strong shear zone, local gouge and well-fractured; minor mudstone apparent. Most intense @ 157.6m.														
			157.7-158.2: Scattered flecks and patches of bright green Cr-mica in medium-grained quartz-eye grit.														
			END OF HOLE = 158.50 metres														

Diam d Drill Log DDH No. **FC-02-06** Date June 14 2

Alteration score 1 - 5

Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
27.8	33.4	QSP	<u>Quartz-Sericite Phyllite</u>; as above (16.9-27.2m); slightly more gritty.		2-3												
			29.0: S2 @ 30 to c.a., orientation turned sharply at fault, dragged?														
			30.4: S2 @ 0-5 to c.a., fractured down centre of core, highly crenulated.														
			33.2: S2 @ 45 to c.a.														
33.4	33.9	Flt	<u>Fault</u>; minor gougy and rubbly fault, likely ~45 to c.a.														
													36.7	36.95	1LF06-22	107	236
33.9	54.0	QSP	<u>Quartz-Sericite Phyllite</u>; generally quite gritty (sand-size) and moderately fractured. Slightly darker reflecting minor carbonaceous content. A few broken up bluish-grey quartz veins, weak to moderate patchy tan carbonate along selvages. Contortions of layering and S2 decreasing. Increased sulphide, mainly pyrite.	tr	5+	tr	tr						38.5	40.2	8287	48	783
													40.2	41.7	8288	20	606
													41.7	43.6	8289	64	943
													43.6	45.1	8290	27	378
			38.5-40.2: Up to 10% sulphides, mainly pyrite, minor sphalerite.		10		min						45.1	46.6	8291	64	931
			41.7-43.6: Bands of disseminated pyrite, minor sphalerite, 5% sulphides.		5		min						46.6	48.5	8292	28	335
			45.7-46.4: Bands of disseminated pyrite, minor sphalerite, 5% sulphides.		5		min						48.5	50.5	8293	40	614
			47.1-47.6: Pebbly, fractured and healed blue-grey quartz vein; not mineralized.										50.5	52.3	8294	39	969
			48.4-49.2: Blue-grey quartz vein, as above. Fracturing increases toward fault.										52.3	52.7	8295	284	2635
			52.3-52.7: Bands of disseminated pyrite, minor sphalerite, 5% sulphides.		5-7		min										
54.0	54.8	Flt	<u>Fault</u> is poorly defined, very poor recovery.														
			53.34-54.25: 25% core recovery.														
			54.25-54.86: 40% core recovery.														
54.8	56.4	QSP	<u>Quartz-Sericite Phyllite</u>; as above(33.9-54.0m), well-sheared, minor gouge.														
56.4	56.7	Flt	<u>Fault</u>; well-defined, strongly gougy fault with 0.5mm quartz eyes in gouge (gritty QSP); 100% core recovery.														
56.7	135.0	QSP	<u>Quartz-Sericite Phyllite</u>; pale grey with local wispy tan Fe-carbonate bands in fine to medium-grained quartz-eye grit (sand-size). Wispy disseminated sulphide mineralization extends to at least 75.0m, forming an identifiable horizon. Generally weakly fractured and competent throughout.														
			56.7-57.6: Well-fractured adjacent to fault, wispy tan carbonate common.														
			57.6-59.4: Disseminated Sulphide Zone														
			57.6-58.5: Increasing bands, 1-10mm thick of wispy, disseminated pyrite.										56.7	58.5	8296	2234	204
			<u>58.5-58.6: Semi-Massive Sulphides</u>; 10cm thick band made up 25% medium-grained pyrite, 5% chalcopyrite (cp), 2% sphalerite (sph), minor galena (gal), and pyrrhotite (po). Very weakly magnetic. Wisps and bands @ 75 to c.a. 1x2cm unmineralized clast in zone. Overall, quite similar to massive sulphide zone in FC-02-05.	<1	25	5	2	<1					58.5	59.4	8297	2507	3978
													58.5	59.44	2LF06-02	2460	1380
													58.5	59.44	3LF06-03	2.08%	9750
													0.45m sample from 0.96m interval?				

Diam[illegible]d Drill Log

DDH No. **FC-02-06**

Date June 14, [illegible]2

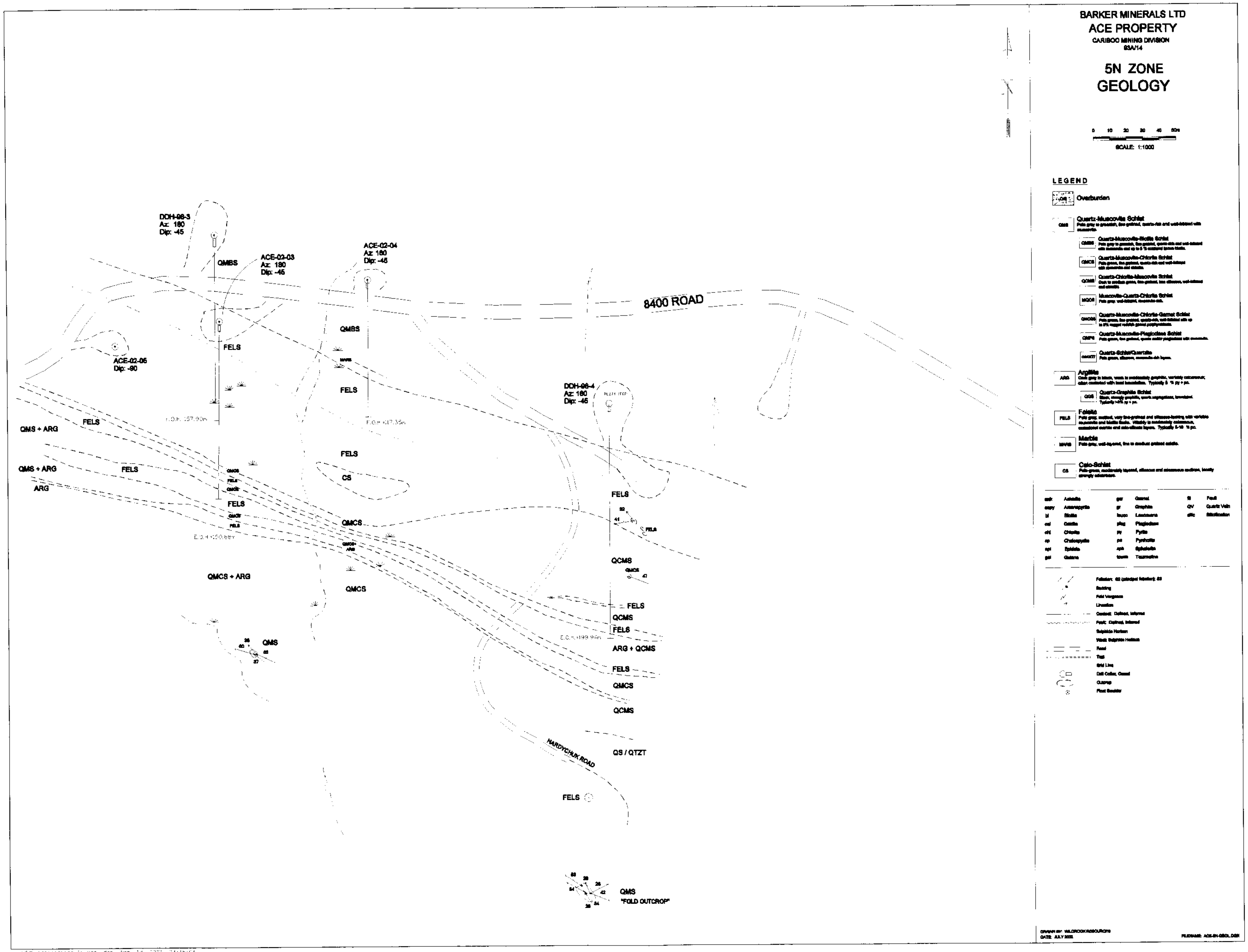
Alteration score 1 - 5

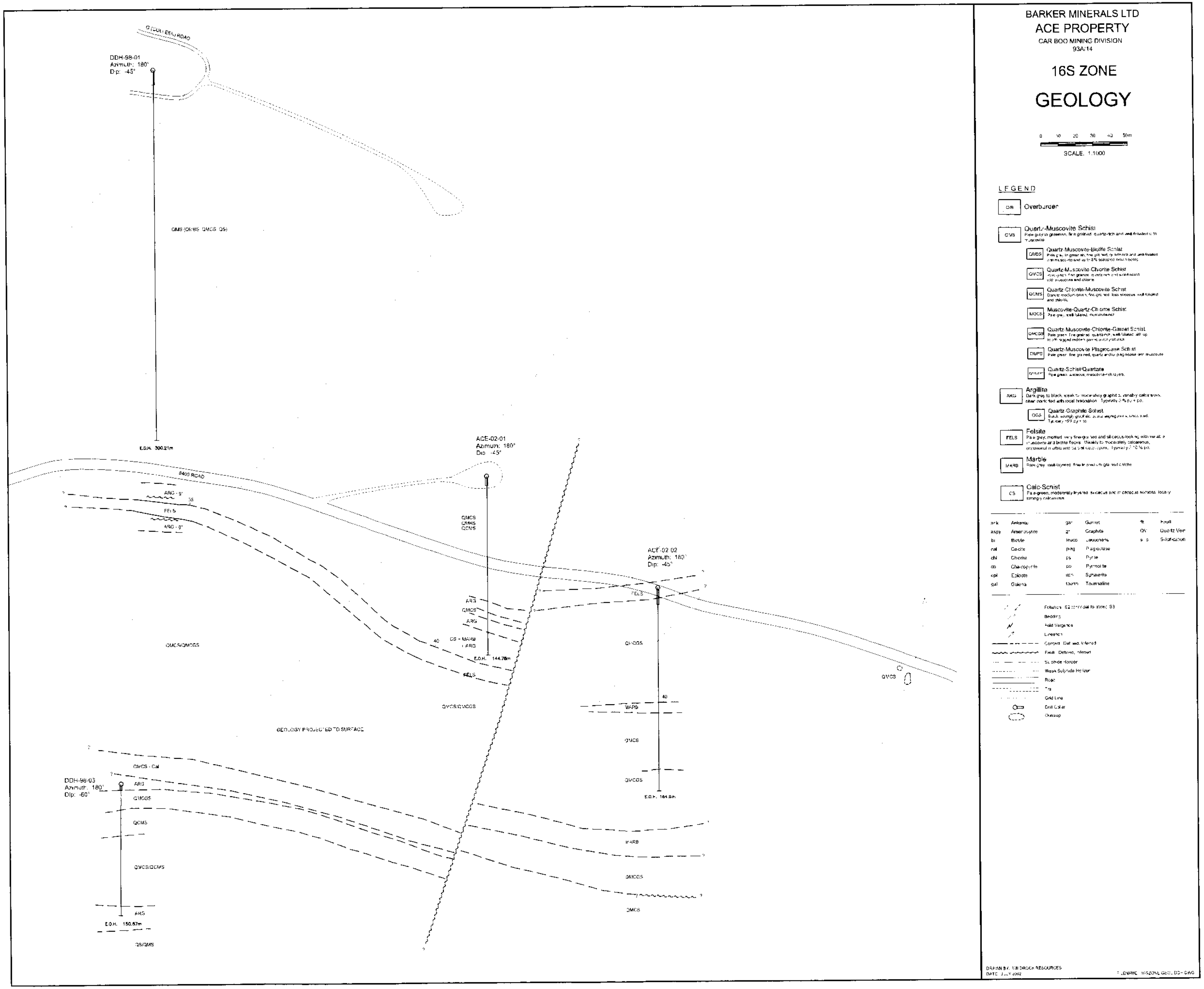
Depth (m) From	To	Rock Code	Description	% Po	% Py	% Cp	% Sph	% Gal	chl	cal	ser	cla	Interval (m) from	to	Sample Number	Cu ppm	Zn ppm
			58.6-58.8: Several bands of disseminated pyrite, minor cp, up to 1cm thick.														
			58.9: 3-4cm layer of semi to massive pyrite, 5% cp.														
			59.1-59.4: Zone of disseminated bands of pyrite, cp, minor sph and galena.														
			59.4: Sharp drop in sulphide content.										59.4	61.2	8298	48	301
			59.9-61.2: Sharp increase in spotted and wispy tan Fe-carbonate, 1-2% pyrite.										61.2	63.2	8299	248	603
			62.2: Band of disseminated pyrite-sph, up to 1cm thick.														
			62.6-63.1: Patchy coarse-grained pyrite, medium-grained po, minor cp & sph.										62.14	62.20	3LF06-36	3390	7880
			63.3-64.9: Weakly fractured, medium-grained grit, excellent fine quartz-eyes in wispy tan carbonate-altered matrix.										62.60	63.09	2LF06-37	1755	618
			64.0: S2 @ 70 to c.a.														
			64.9-75.2: Disseminated stringer sulphide zone, 5-7% mainly pyrite, minor sph, cp, po.										64.9	67.0	8300	1066	225
			65.5: 10cm white quartz vein, 5% po, minor cp, patchy and fracture-controlled tan carbonate, possible black hairline tourmaline.										67.0	69.0	8301	2278	317
													69.0	71.0	8302	426	359
			65.8-66.2: Coarse and fine-grained wispy pyrite, minor po & sph.										71.0	73.0	8303	173	443
			66.3: White vuggy quartz vein, 6-7cm thick @ 60 to c.a.										73.0	75.2	8304	175	776
			66.4-66.8: Wispy dissmeinated py, minor cp. sph, po.														
			67.7-68.4: Strong zone of disseminated sulphides, mainly pyrite; strong cp @ 68.2m.										67.00	69.00	2LF06-04	1480	236
			68.4-75.2: Decreasing sulphides, less frequent mineralized "pulses".										67.75	67.80	3LF06-05	1.27%	842
			72.0-72.5: Zone of strong wispy tan Fe-carbonate, minor pyrite, more siliceous.										68.20	68.35	3LF06-06	7840	1150
			74.3-74.7: Wispy pyrite, sph, rare galena.														
			75.0-75.2: Py-po, trace sph-gal associated with irregular quartz veining; patchy tan Fe-carbonate common.										71.9	72.3	3LF06-41	512	468
			75.2-135.0: Weaker sulphides but increased tan carbonate flooding in wispy patches and spots. Continuing only weakly fractured, little brittle shearing.		2												
			83.7-135.0: Increased coarse-grained pyrite; very consistent unit.		2												
			84.9: 2-3cm gougy shear, little associated fracturing; S2 @ 75-80 to c.a.														
			86.8: Blebby py-cp, po, minor quartz with cp, many fine Fe-carb spots.										86.6	86.8	1LF06-07	1560	380
			89.1-135.0: Developing breccia texture with bullet-shaped lapilli-size clasts in slightly darker matrix; much more subtle texture than in FC-02-05. Unit is still quartz-eye grit locally coarser and packed.										86.87	87.20	1LF06-08	991	476
			90.5: 3cm vein of Fe-carb, strong Cr-mica, brighter green; wispy selvages @ 80 to c.a.														
			93.2, 93.5: Similar zones of wispy Fe-carb and associated Cr-mica; 4 & 15cm thick, respectively.														
			99.4: Quartz pebble grit, very poorly sorted with wispy Fe-carb in matrix.														
			99.5-99.7: Packed quartz-eyes, well-sorted with average diameter of 1-2mm, in grit.										99.4	99.7	1LF06-10	27	46
			101.5-104.3: Breccia textures coincident with increased fracturing and a few partially healed blue-grey quartz veins.														
			104.3-108.8: Similar to above but less intense and obvious.														

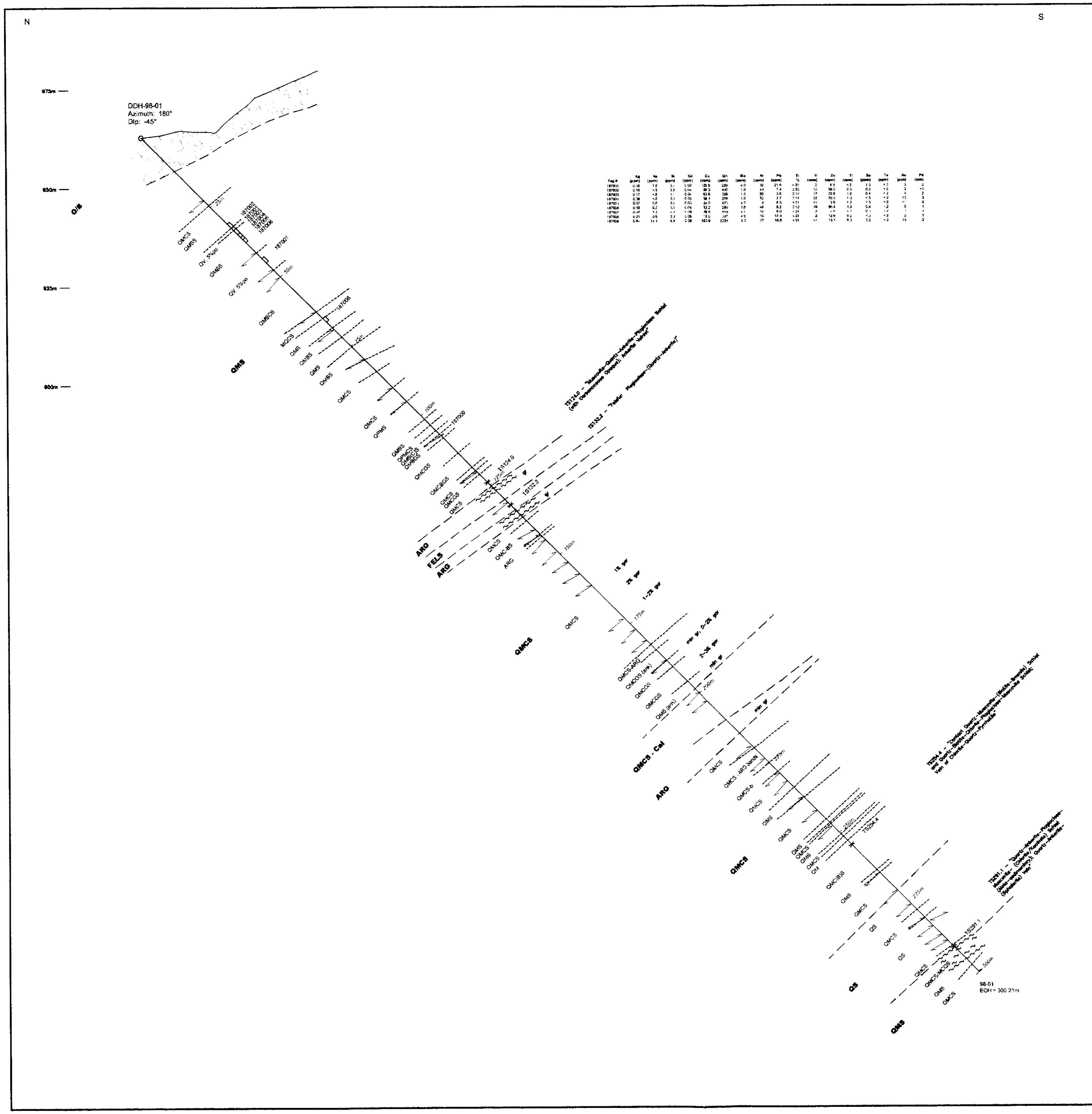
Diamond Drill Log DDH No. **FC-02-06** Date June 14, [illegible]2

Alteration score 1 - 5

Depth (m)		Rock	Description	%	%	%	%	%	chl	cal	ser	cla	Interval (m)		Sample	Cu	Zn
From	To	Code		Po	Py	Cp	Sph	Gal					from	to	Number	ppm	ppm
			116.3, 116.6, 118.2, 118.5: Sections of packed, coarser quartz-eye grit, 5-8cm thick; @ 70-80 to c.a.										112.0	112.3	1LF06-11	22	50
			118.6-122.1: Coarse breccia with unstrained but gritty pale grey sections.														
			122.1-123.4: Slightly more phyllitic, S2 @ 80 to c.a., variable.														
			123.4: Blue-grey quartz vein, 8cm thick, fractured but healed and mottled appearance; late coarse-grained cubic pyrite. Pyrite continues to 124.2m.														
			124.6-124.9: Well-fractured along blue-grey quartz vein.														
			125.7-125.9: Minor fault, gouge and rubble.														
			128.9-130.4: Slightly darker, more phyllitic.														
			130.4-134.1: Weak breccia texture, coarse clasts of fine to moderately gritty, moderate Fe-carb, wispy and tan coloured through matrix.														
			134.1-134.4: Fractured white quartz vein in grit, well fractured, no sulphides.														
			134.4-135.0: Coarse more pebbly-looking grit, quartz grains 2-5mm, aligned, clast-supported, better sorted in sericitic matrix.														
			END OF HOLE = 135.03 metres														


BARKER MINERALS LTD
ACE PROPERTY
CARIBOO MINING DIVISION
93A/14
5N ZONE
GEOLOGY
0 10 20 30 40 50m
SCALE: 1:1000
LEGEND
OB Overburden
QMS Quartz-Muscovite Schist
QMBS Quartz-Muscovite-Biotite Schist
QMCS Quartz-Muscovite-Chlorite Schist
QCMS Quartz-Chlorite-Muscovite Schist
MQCS Muscovite-Quartz-Chlorite Schist
QMCGS Quartz-Muscovite-Chlorite-Garnet Schist
QMPS Quartz-Muscovite-Plagioclase Schist
QS/QTZT Quartz-Schist/Quartzite
ARG Argillite
QGS Quartz-Graphite Schist
FELS Felsite
MARB Marble
CS Calc-Schist
ank Ankerite
aspy Arsenopyrite
bi Biotite
cal Calcite
gar Garnet
gr Graphite
leuco Leucoxene
plag Plagioclase
py Pyrite
po Pyrrhotite
sph Sphalerite
tourm Tourmaline
flt Fault
QV Quartz Vein
silic Silicification
DDH-98-3
Az: 180
Dip: -45
ACE-02-03
Az: 180
Dip: -45
ACE-02-04
Az: 180
Dip: -45
ACE-02-05
Dip: -90
DDH-98-4
Az: 180
Dip: -45
8400 ROAD
HARDYCHUK ROAD
QMBS
FELS
QMS + ARG
ARG
QMCS
CS
QMCS + ARG
QMS
QCMS
ARG + QCMS
QS / QTZT
QMS
"FOLD OUTCROP"


BARKER MINERALS LTD
ACE PROPERTY
CAR BOO MINING DIVISION
93A/14
16S ZONE
GEOLOGY
0 10 20 30 40 50m
SCALE 1:1000
LEGEND
O/B Overburden
QMS Quartz-Muscovite Schist
QMBS Quartz-Muscovite-Biotite Schist
QMCS Quartz-Muscovite-Chlorite Schist
QCMS Quartz-Chlorite-Muscovite Schist
MQCS Muscovite-Quartz-Chlorite Schist
QMCGS Quartz-Muscovite-Chlorite-Garnet Schist
QMPS Quartz-Muscovite Plagioclase Schist
Quartz-Schist/Quartzite
ARG Argillite
QGS Quartz-Graphite Schist
FELS Felsite
MARB Marble
CS Calc-Schist
ank Ankerite
aspy Arsenopyrite
bi Biotite
cal Calcite
chl Chlorite
cp Chalcopyrite
epi Epidote
gal Galena
gar Garnet
gr Graphite
leuco Leucoxene
plag Plagioclase
py Pyrite
po Pyrrhotite
sph Sphalerite
tourm Tourmaline
flt Fault
QV Quartz Vein
Foliation: S2 (principal foliation); S3
Bedding
Fold Vergence
Lineation
Contact: Defined, Inferred
Fault: Defined, Inferred
Sulphide Horizon
Weak Sulphide Horizon
Road
Trail
Grid Line
Drill Collar
Outcrop
DDH-98-01
Azimuth: 180°
Dip: -45°
E.O.H. 300.21m
ACE-02-01
Azimuth: 180°
Dip: -45°
E.O.H. 144.78m
ACE-02-02
Azimuth: 180°
Dip: -45°
DDH-98-03
Azimuth: 180°
Dip: -60°
E.O.H. 150.87m
8400 ROAD
GEOLOGY PROJECTED TO SURFACE
QMCS/QMCGS
QMCS
QMBS
QCMS
ARG
FELS
MARB
QMCGS
QVCS



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187001	0.30	1.8	[illegible]	0.03	122.8	[illegible]	4.0	32	21.6	<.01	2	8.9	<.2	1.3	<2	3	2
187002	0.10	<.5	3.5	0.04	89.3	400	1.0	44	7.4	0.03	12	58.0	0.3	0.3	<2	3	<1
187003	0.17	<.5	[illegible]	0.04	63.6	309	1.3	90	5.6	0.10	17	72.8	1.0	0.4	<2	4	2
187004	0.39	<.5	3.1	0.02	38.4	209	1.0	52	3.7	0.14	22	76.5	1.3	<.3	<2	10	3
187005	0.07	0.8	3.4	0.03	24.0	475	4.7	4	8.5	<.01	<1	3.9	<.2	<.3	<2	[illegible]	4
187006	0.10	0.7	[illegible]	0.04	53.2	290	1.8	46	8.2	0.12	19	86.4	1.3	0.4	<2	3	1
187007	0.07	1.3	[illegible]	[illegible]	[illegible]	[illegible]	4.1	[illegible]	6.0	<.01	2	[illegible]	<.2	[illegible]	<2	[illegible]	[illegible]
187008	0.21	0.6	3.3	0.06	[illegible]	307	4.0	10	17.4	<.01	2	12.6	0.2	<.3	<2	2	1
187009	[illegible]	14.1	6.9	0.09	[illegible]	2254	3.3	27	18.8	<.01	<1	15.1	0.3	2.2	<2	25	3

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION

93A/14

16S ZONE
DDH-98-01 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

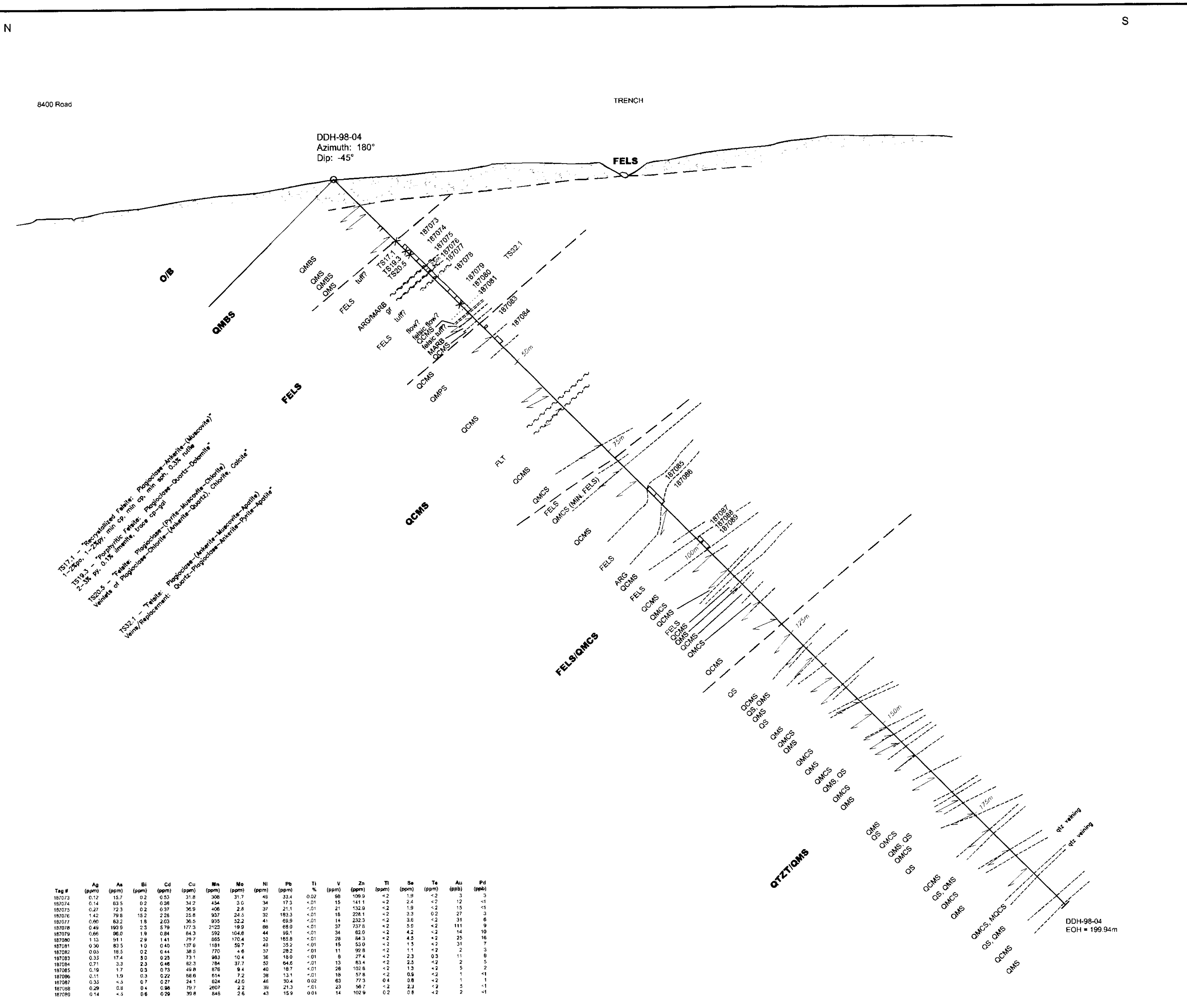
- O/B — Overburden
- QMS — Quartz-Muscovite Schist. Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite.
  - QMBS — Quartz-Muscovite-Biotite Schist
  - QMCS — Quartz-Muscovite-Chlorite Schist
  - QCMS — Quartz-Chlorite-Muscovite Schist
  - MQCS — Muscovite-Quartz-Chlorite Schist
  - QMCGS — Quartz-Muscovite-Chlorite-Garnet Schist
  - QMPS — Quartz-Muscovite-Plagioclase Schist
  - QS/QTZ — Quartz-Schist/Quartzite
- ARG — Argillite. Dark grey to black, weak to moderately graphitic, variably calcareous, often contorted with local brecciation. Typically 3 % py + po.
  - QGS — Quartz-Graphite Schist
- FELS — Felsite
- MARB — Marble. Pale grey, well-layered, fine to medium grained calcite.
- CS — Calc-Schist. Pale green, moderately layered, siliceous and micaceous sections, locally strongly calcareous.

ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bi	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
epi	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

- Foliation: S2 (principal foliation); S3
- Bedding
- Fold Vergence
- Lineation
- Contact: Defined, Inferred
- Fault: Defined, Inferred
- Sulphide Horizon
- Weak Sulphide Horizon

DRAWN BY: SOLDROCK RESOURCES
DATE: JULY 2002

FILENAME: DDH-98-01.DWG



Tag #	Ag (ppm)	As (ppm)	Bi (ppm)	Cd (ppm)	Cu (ppm)	Mn (ppm)	Mo (ppm)	Ni (ppm)	Pb (ppm)	Ti %	V (ppm)	Zn (ppm)	Tl (ppm)	Se (ppm)	Te (ppm)	Au (ppb)	Pd (ppb)
187073	0.12	15.7	0.2	0.53	31.8	308	31.7	49	33.4	0.02	98	109.9	<.2	1.9	<.2	3	3
187074	0.14	63.5	0.2	0.36	34.2	454	3.0	34	17.3	<.01	15	141.1	<.2	2.4	<.2	12	<1
187075	0.27	72.3	0.2	0.37	36.9	408	2.8	37	21.1	<.01	21	132.9	<.2	1.9	<.2	15	<1
187076	1.42	79.8	15.2	2.28	25.8	937	24.5	32	183.3	<.01	18	228.1	<.2	3.3	0.2	27	3
187077	0.60	63.2	1.8	2.03	36.5	935	52.2	41	69.9	<.01	14	232.3	<.2	3.6	<.2	31	6
187078	0.49	193.9	2.3	5.79	177.3	2123	19.9	66	68.0	<.01	37	737.8	<.2	5.9	<.2	111	9
187079	0.66	96.0	1.9	0.84	64.3	592	104.6	44	99.1	<.01	34	62.0	<.2	4.2	<.2	14	10
187080	1.13	91.1	2.9	1.41	79.7	865	170.4	52	165.8	<.01	28	84.3	<.2	4.5	<.2	25	16
187081	0.30	63.5	1.0	0.40	137.6	1181	59.7	49	35.2	<.01	15	53.0	<.2	1.5	<.2	31	7
187082	0.05	16.5	0.2	0.44	38.5	770	4.6	37	28.2	<.01	11	92.8	<.2	1.1	<.2	2	3
187083	0.33	17.4	5.0	0.25	73.1	983	10.4	36	18.0	<.01	8	27.4	<.2	2.3	0.3	11	8
187084	0.71	3.3	2.3	0.46	62.3	784	37.7	52	64.6	<.01	13	63.4	<.2	2.5	<.2	2	5
187085	0.19	1.7	0.3	0.73	49.8	876	9.4	40	18.7	<.01	28	102.8	<.2	1.3	<.2	5	2
187086	0.11	1.9	0.3	0.22	68.6	614	7.2	38	13.1	<.01	18	57.8	<.2	0.9	<.2	1	<1
187087	0.33	<.5	0.7	0.27	24.1	624	42.0	46	30.4	0.02	63	77.3	0.4	0.8	<.2	1	1
187088	0.29	0.8	0.4	0.98	79.7	2607	2.2	39	21.3	<.01	23	56.7	<.2	2.3	<.2	5	<1
187089	0.14	<.5	0.6	0.29	39.8	846	2.6	43	15.9	0.01	14	102.9	0.2	0.6	<.2	2	<1

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION
93A/14

5N ZONE
DDH-98-04 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

O/B Overburden

QMS **Quartz-Muscovite Schist**
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite.

QMBS **Quartz-Muscovite-Biotite Schist**
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite and up to 5% scattered brown biotite.

QMCS **Quartz-Muscovite-Chlorite Schist**
Pale green, fine grained, quartz-rich and well-foliated with muscovite and chlorite.

QCMS **Quartz-Chlorite-Muscovite Schist**
Dark to medium green, fine-grained, less siliceous, well-foliated and chloritic.

MQCS **Muscovite-Quartz-Chlorite Schist**
Pale grey, well-foliated, muscovite-rich.

QMCGS **Quartz-Muscovite-Chlorite-Garnet Schist**
Pale green, fine grained, quartz-rich, well-foliated with up to 5% ragged reddish garnet porphyroblasts.

QMPS **Quartz-Muscovite-Plagioclase Schist**
Pale green, fine grained, quartz and/or plagioclase with muscovite.

QS/QTZT **Quartz-Schist/Quartzite**
Pale green, siliceous, muscovite-rich layers.

ARG **Argillite**
Dark grey to black, weak to moderately graphitic, variably calcareous; often contorted with local brecciation. Typically 5% py + po.

QGS **Quartz-Graphite Schist**
Black, strongly graphitic, quartz segregations, brecciated. Typically >5% py + po.

FELS **Felsite**
Pale grey, mottled, very fine-grained and siliceous-looking with variable muscovite and biotite flecks. Weakly to moderately calcareous, occasional marble and calc-silicate layers. Typically 5-10% po.

MARB **Marble**
Pale grey, well-layered, fine to medium grained calcite.

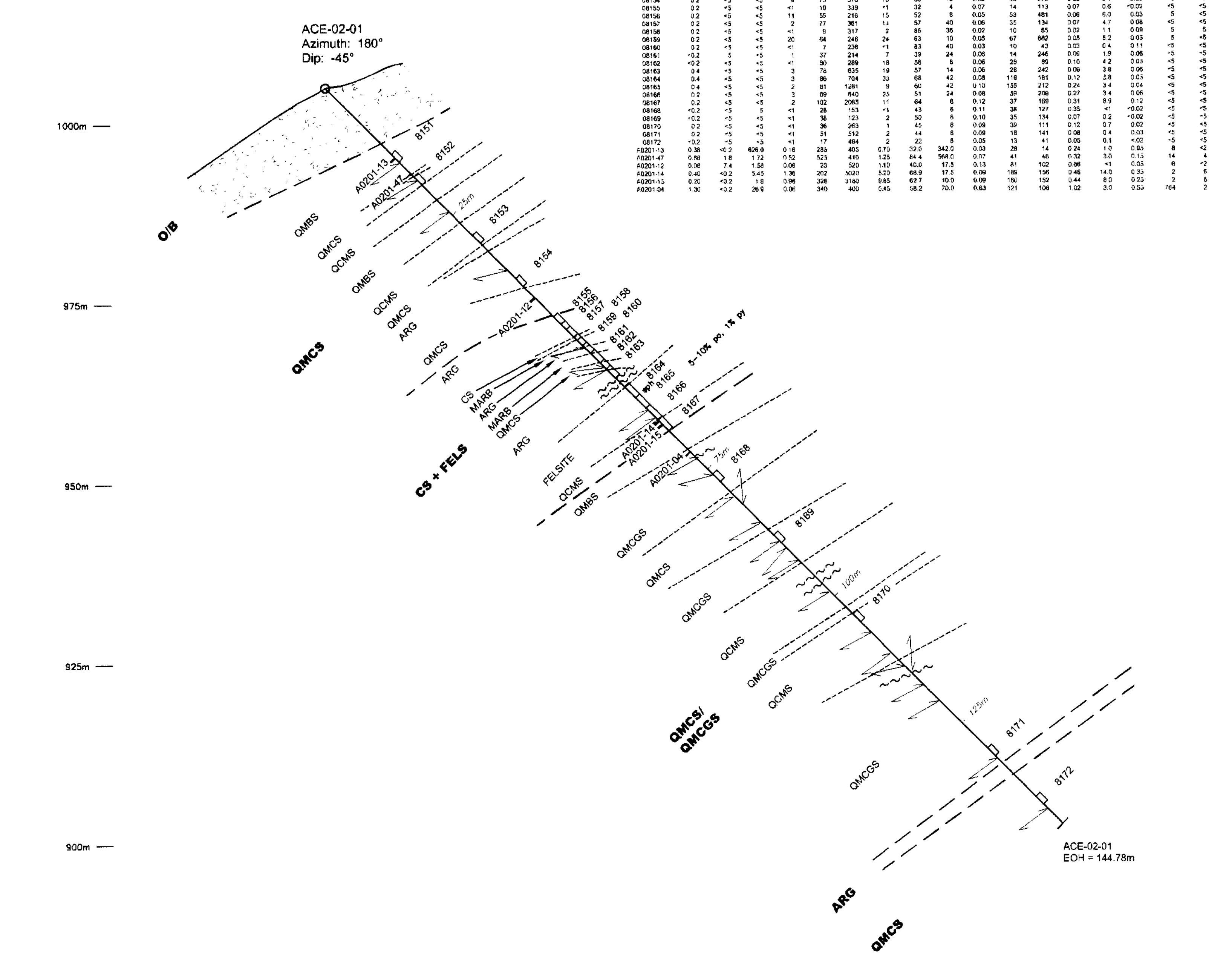
CS **Calc-Schist**
Pale-green, moderately layered, siliceous and micaceous sections, locally strongly calcareous.

ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bi	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
epi	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

Foliation: S2 (principal foliation); S3
Bedding
Fold Vergence
Lineation
Contact: Defined, Inferred
Fault: Defined, Inferred
Sulphide Horizon
Weak Sulphide Horizon

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

FILENAME: DDH-98-04.DWG



Tag #	Ag (ppm)	As (ppm)	Bi (ppm)	Cd (ppm)	Cu (ppm)	Mn (ppm)	Mo (ppm)	Ni (ppm)	Pb (ppm)	Ti %	V (ppm)	Zn (ppm)	Tl (ppm)	Se (ppm)	Te (ppm)	Au (ppb)	Pd (ppb)
08151	0.2	<5	140	<1	57	345	2	28	48	0.13	17	58	0.50	0.3	0.07	5	<5
08152	<0.2	<5	<5	<1	22	245	<1	33	4	0.09	15	76	0.32	0.3	<0.02	<5	<5
08153	0.4	<5	<5	1	34	347	2	40	6	0.07	25	124	0.06	1.1	<0.02	<5	<5
08154	0.2	<5	<5	4	75	578	16	50	10	0.06	40	278	0.08	5.1	0.03	5	<5
08155	0.2	<5	<5	<1	19	339	<1	32	4	0.07	14	113	0.07	0.6	<0.02	<5	<5
08156	0.2	<5	<5	11	55	216	15	52	8	0.05	53	481	0.08	6.0	0.03	5	<5
08157	0.2	<5	<5	2	77	381	14	57	40	0.06	35	134	0.07	4.7	0.08	<5	<5
08158	0.2	<5	<5	<1	9	317	2	86	35	0.02	10	65	0.02	1.1	0.09	5	5
08159	0.2	<5	<5	20	64	248	24	63	10	0.05	67	682	0.05	5.2	0.05	5	<5
08160	0.2	<5	<5	<1	7	236	<1	83	40	0.03	10	43	0.03	0.4	0.11	<5	<5
08161	<0.2	5	<5	1	37	214	7	39	24	0.06	14	246	0.09	1.9	0.06	<5	<5
08162	<0.2	<5	<5	<1	80	289	18	58	8	0.06	29	89	0.10	4.2	0.05	<5	<5
08163	0.4	<5	<5	3	78	635	19	57	14	0.06	28	242	0.09	3.8	0.06	<5	<5
08164	0.4	<5	<5	3	86	704	33	68	42	0.08	118	181	0.12	3.8	0.05	<5	<5
08165	0.4	<5	<5	2	81	1281	9	60	42	0.10	155	212	0.24	3.4	0.04	<5	<5
08166	0.2	<5	<5	3	69	640	25	51	24	0.08	59	209	0.27	3.4	0.06	<5	<5
08167	0.2	<5	<5	2	102	2065	11	64	6	0.12	37	169	0.31	8.9	0.12	<5	<5
08168	<0.2	<5	5	<1	28	153	<1	43	6	0.11	38	127	0.35	<1	<0.02	<5	<5
08169	<0.2	<5	<5	<1	38	123	2	50	6	0.10	35	134	0.07	0.2	<0.02	<5	<5
08170	0.2	<5	<5	<1	36	263	1	45	8	0.09	39	111	0.12	0.7	0.02	<5	<5
08171	0.2	<5	<5	<1	51	512	2	44	6	0.09	18	141	0.08	0.4	0.03	<5	<5
08172	<0.2	<5	<5	<1	17	484	2	22	8	0.05	13	41	0.05	0.1	<.02	<5	<5
A0201-13	0.38	<0.2	626.0	0.16	285	405	0.70	32.0	342.0	0.03	28	14	0.24	1.0	0.95	8	<2
A0201-47	0.88	1.8	1.72	0.52	525	410	1.25	84.4	568.0	0.07	41	46	0.32	3.0	0.15	14	4
A0201-12	0.08	7.4	1.58	0.08	23	520	1.10	40.0	17.5	0.13	81	102	0.86	<1	0.05	6	<2
A0201-14	0.40	<0.2	5.45	1.36	202	5020	5.20	68.9	17.5	0.09	169	156	0.46	14.0	0.35	2	6
A0201-15	0.20	<0.2	1.8	0.96	328	3180	9.85	62.7	10.0	0.09	160	152	0.44	8.0	0.25	2	6
A0201-04	1.30	<0.2	26.9	0.06	340	400	0.45	58.2	70.0	0.63	121	106	1.02	3.0	0.55	764	2

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION
93A/14

16S ZONE
ACE-02-01 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

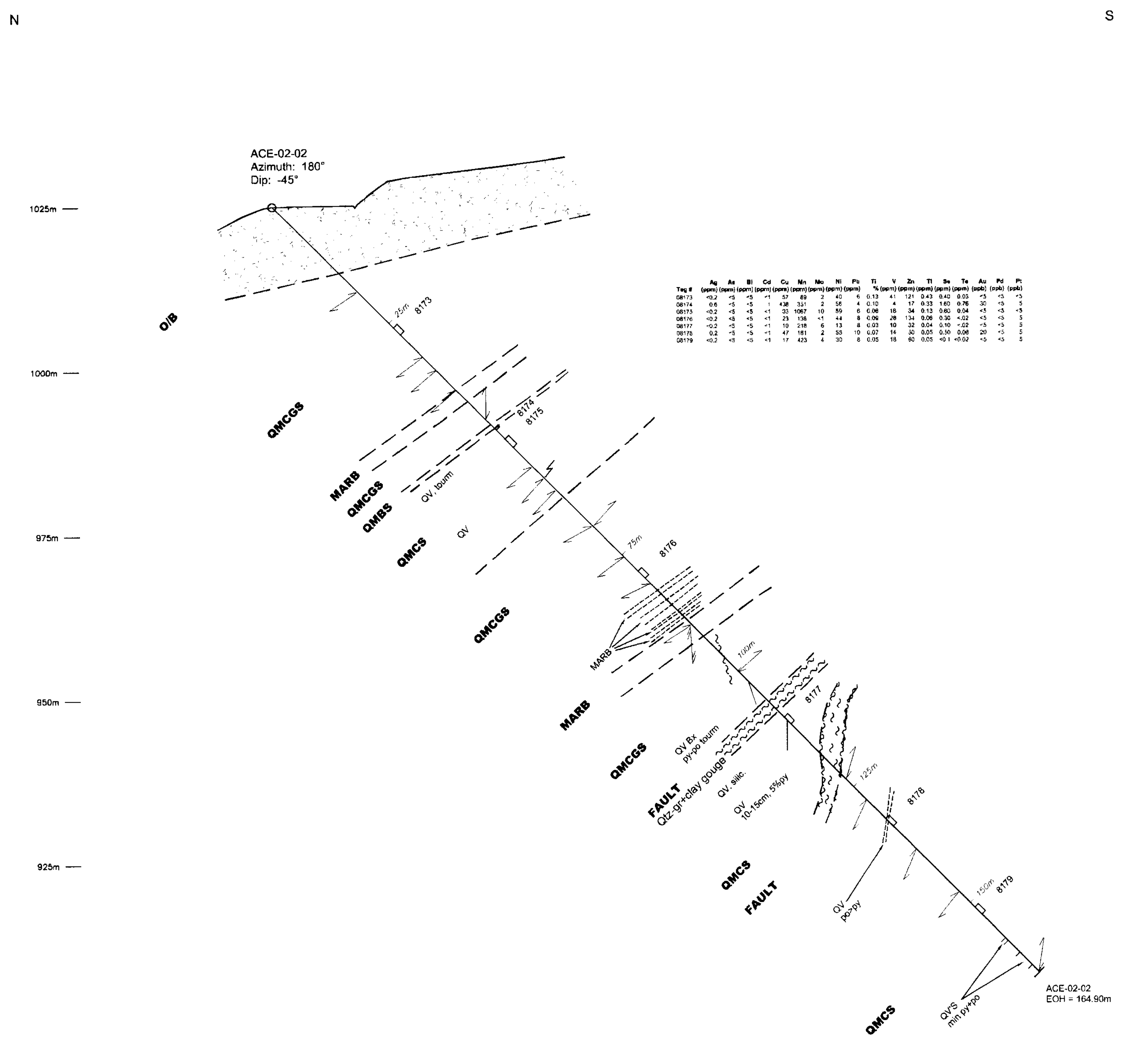
O/B	Overburden
QMS	**Quartz-Muscovite Schist** Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite.
QMBS	**Quartz-Muscovite-Biotite Schist** Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite and up to 5% scattered brown biotite.
QMCS	**Quartz-Muscovite-Chlorite Schist** Pale green, fine grained, quartz-rich and well-foliated with muscovite and chlorite.
QCMS	**Quartz-Chlorite-Muscovite Schist** Dark to medium green, fine-grained, less siliceous, well-foliated and chloritic.
MQCS	**Muscovite-Quartz-Chlorite Schist** Pale grey, well-foliated, muscovite-rich.
QMCGS	**Quartz-Muscovite-Chlorite-Garnet Schist** Pale green, fine grained, quartz-rich, well-foliated with up to 5% ragged reddish garnet porphyroblasts.
QMPS	**Quartz-Muscovite-Plagioclase Schist** Pale green, fine grained, quartz and/or plagioclase with muscovite.
QS/QTZT	**Quartz-Schist/Quartzite** Pale green, siliceous, muscovite-rich layers.
ARG	**Argillite** Dark grey to black, weak to moderately graphitic, variably calcareous; often contorted with local brecciation. Typically 5% py + po.
QGS	**Quartz-Graphite Schist** Black, strongly graphitic, quartz segregations, brecciated. Typically >5% py + po.
FELS	**Felsite** Pale grey, mottled, very fine-grained and siliceous-looking with variable muscovite and biotite flecks. Weakly to moderately calcareous, occasional marble and calc-silicate layers. Typically 5-10% po.
MARB	**Marble** Pale grey, well-layered, fine to medium grained calcite.
CS	**Calc-Schist** Pale-green, moderately layered, siliceous and micaceous sections, locally strongly calcareous.

ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bi	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
epi	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

Foliation: S2 (principal foliation); S3
Bedding
Fold Vergence
Lineation
Contact: Defined, Inferred
Fault: Defined, Inferred
Sulphide Horizon
Weak Sulphide Horizon

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

FILENAME: DDH-ACE-02-01.DWG



Tag #	Ag (ppm)	As (ppm)	Bi (ppm)	Cd (ppm)	Cu (ppm)	Mn (ppm)	Mo (ppm)	Ni (ppm)	Pb (ppm)	Ti %	V (ppm)	Zn (ppm)	Tl (ppm)	Se (ppm)	Te (ppm)	Au (ppb)	Pd (ppb)	Pt (ppb)
08173	<0.2	<5	<5	<1	57	89	2	40	6	0.13	41	121	0.43	0.40	0.03	<5	<5	<5
08174	0.6	<5	<5	1	438	351	2	56	4	0.10	4	17	0.33	1.60	0.76	30	<5	5
08173	<0.2	<5	<5	<1	33	1087	10	59	6	0.08	18	34	0.13	0.60	0.04	<5	<5	<5
08176	<0.2	<5	<5	<1	23	138	<1	44	8	0.09	28	134	0.06	0.30	<.02	<5	<5	5
08177	<0.2	<5	<5	<1	10	218	6	13	8	0.03	10	32	0.04	0.10	<.02	<5	<5	5
08178	0.2	<5	<5	<1	47	161	2	53	10	0.07	14	30	0.05	0.50	0.08	20	<5	5
08179	<0.2	<5	<5	<1	17	423	4	30	8	0.05	18	60	0.05	<0.1	<0.02	<5	<5	5

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION
93A/14

16S ZONE
ACE-02-02 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

O/B Overburden

QMS Quartz-Muscovite Schist
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite.

- **QMBS** Quartz-Muscovite-Biotite Schist
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite and up to 5% scattered brown biotite.
- **QMCS** Quartz-Muscovite-Chlorite Schist
Pale green, fine grained, quartz-rich and well-foliated with muscovite and chlorite.
- **QCMS** Quartz-Chlorite-Muscovite Schist
Dark to medium green, fine-grained, less siliceous, well-foliated and chlontic.
- **MQCS** Muscovite-Quartz-Chlorite Schist
Pale grey, well-foliated, muscovite-rich.
- **QMCGS** Quartz-Muscovite-Chlorite-Garnet Schist
Pale green, fine grained, quartz-rich, well-foliated with up to 5% ragged reddish garnet porphyroblasts.
- **QMPS** Quartz-Muscovite-Plagioclase Schist
Pale green, fine grained, quartz and/or plagioclase with muscovite.
- **QS/QTZT** Quartz-Schist/Quartzite
Pale green, siliceous, muscovite-rich layers.

ARG Argillite
Dark grey to black, weak to moderately graphitic, variably calcareous; often contorted with local brecciation. Typically 5% py + po.

- **QGS** Quartz-Graphite Schist
Black, strongly graphitic, quartz segregations, brecciated. Typically >5% py + po.

FELS Felsite
Pale grey, mottled, very fine-grained and siliceous-looking with variable muscovite and biotite flecks. Weakly to moderately calcareous, occasional marble and calc-silicate layers. Typically 5-10% po.

MARB Marble
Pale grey, well-layered, fine to medium grained calcite.

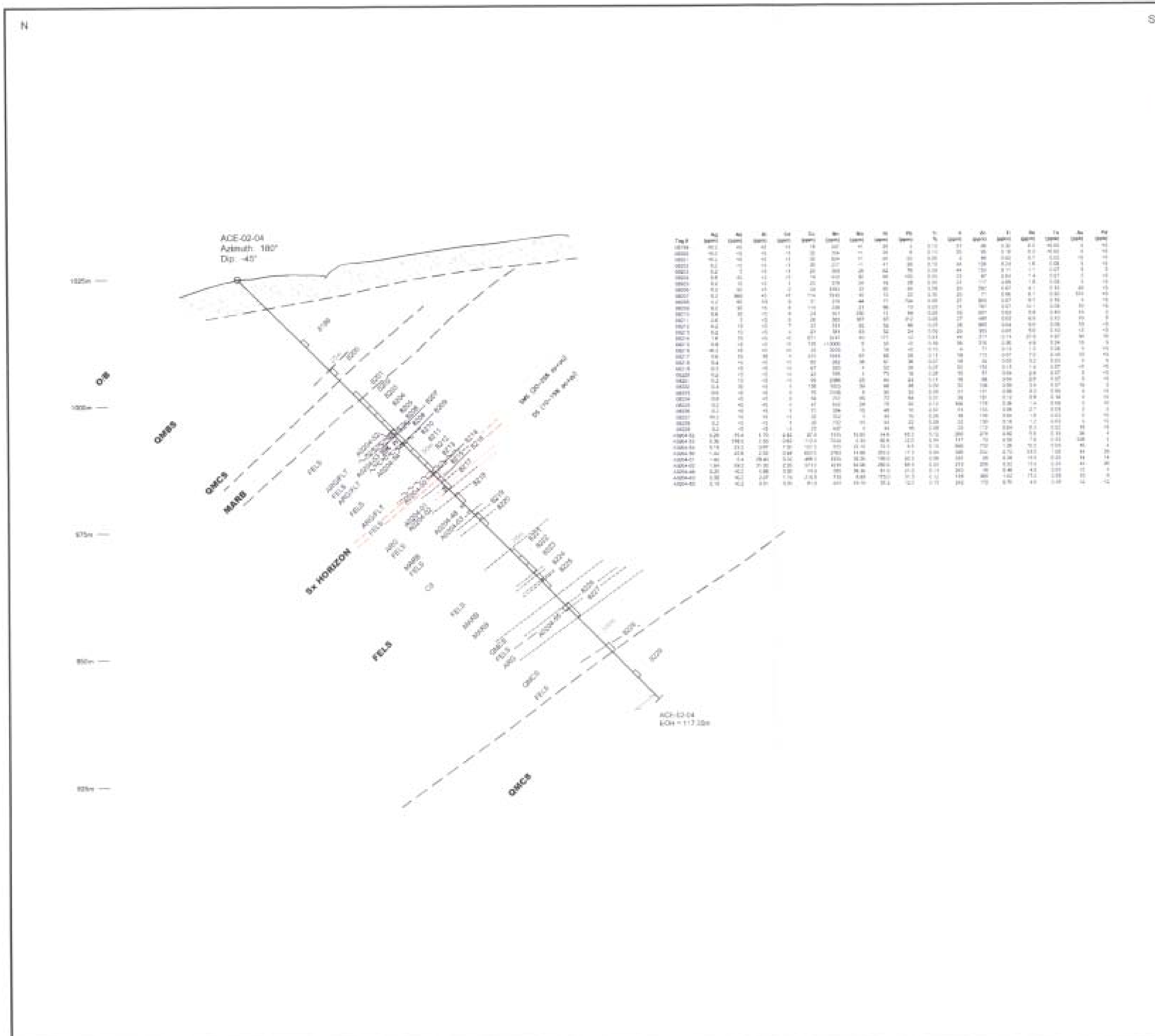
CS Calc-Schist
Pale-green, moderately layered, siliceous and micaceous sections, locally strongly calcareous.

ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bl	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
epi	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

Foliation: S2 (principal foliation); S3
Bedding
Fold Vergence
Lineation
Contact: Defined, Inferred
Fault: Defined, Inferred
Sulphide Horizon
Weak Sulphide Horizon

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

FILENAME: DDH-ACE-02-02.DWG



BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION

93A/14

5N ZONE

ACE-02-04 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

- O/B — Overburden
- QMS — Quartz-Muscovite Schist
  - QMBS — Quartz-Muscovite-Biotite Schist
  - QMCS — Quartz-Muscovite-Chlorite Schist
  - QCMS — Quartz-Chlorite-Muscovite Schist
  - MQCS — Muscovite-Quartz-Chlorite Schist
  - QMCGS — Quartz-Muscovite-Chlorite-Garnet Schist
  - QMPS — Quartz-Muscovite-Plagioclase Schist
  - QS/QTZT — Quartz-Schist/Quartzite
- ARG — Argillite
  - QGS — Quartz-Graphite Schist
- FELS — Felsite
- MARB — Marble
- CS — Calc-Schist

ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bi	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
ep	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

Contact: Defined, Inferred

Fault: Defined, Inferred

Sulphide Horizon

DRAWN BY: WILDROCK RESOURCES

DATE: JULY 2002

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION

5N ZONE
ACE-02-03 &
DDH-98-03 SECTION

VIEWING EAST

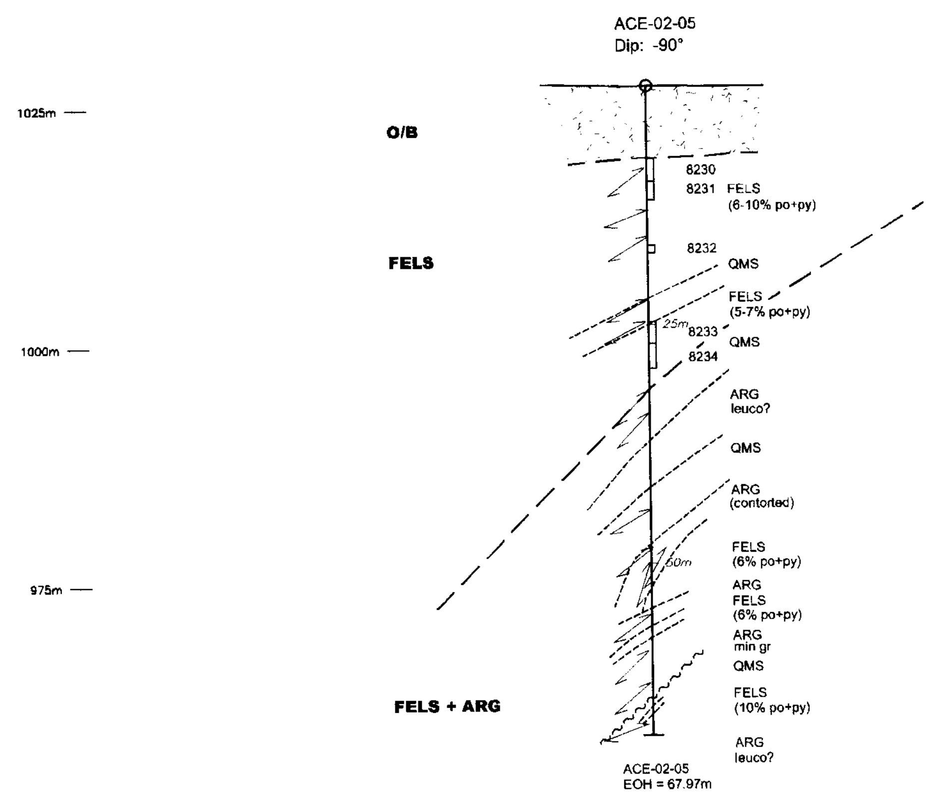
SCALE 1:500

LEGEND

- Overburden
- Quartz-Muscovite Schist
  - Quartz-Muscovite-Biotite Schist
  - Quartz-Muscovite-Chlorite Schist
  - Quartz-Chlorite-Muscovite Schist
  - Muscovite-Quartz-Chlorite Schist
  - Quartz-Muscovite-Chlorite-Garnet Schist
  - Quartz-Muscovite-Plagioclase Schist
  - Quartz-Schist/Quartzite
- Argillite
  - Quartz-Graphite Schist
- Felsite
- Marble
- Calc-Schist

N

S



Tag #	Ag (ppm)	As (ppm)	Bi (ppm)	Cd (ppm)	Cu (ppm)	Mn (ppm)	Mo (ppm)	Ni (ppm)	Pb (ppm)	Ti %	V (ppm)	Zn (ppm)	Tl (ppm)	Se (ppm)	Te (ppm)	Au (ppb)	Pd (ppb)
06230	0.2	<5	<5	11	66	667	45	60	34	0.08	42	385	0.12	4.00	0.07	5	<5
06231	0.2	<5	<5	1	122	2168	15	66	22	0.13	93	158	0.23	4.20	0.10	<5	<5
06232	<0.2	<5	<5	<1	17	504	7	11	<2	0.05	7	48	0.03	0.30	<0.02	<5	<5
06233	0.2	<5	<5	<1	50	468	<1	29	6	0.08	15	57	0.08	2.60	0.04	<5	<5
06234	0.4	<5	<5	<1	60	584	<1	48	20	0.10	63	157	0.21	2.50	0.04	<5	<5

BARKER MINERALS LTD

ACE PROPERTY

CARIBOO MINING DIVISION
93A/14

5N ZONE
ACE-02-05 SECTION

VIEWING EAST

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

O/B — Overburden

QMS — **Quartz-Muscovite Schist**
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite.

QMBS — **Quartz-Muscovite-Biotite Schist**
Pale grey to greenish, fine grained, quartz-rich and well-foliated with muscovite and up to 5% scattered brown biotite.

QMCS — **Quartz-Muscovite-Chlorite Schist**
Pale green, fine grained, quartz-rich and well-foliated with muscovite and chlorite

QCMS — **Quartz-Chlorite-Muscovite Schist**
Dark to medium green, fine-grained, less siliceous, well-foliated and chloritic.

MQCS — **Muscovite-Quartz-Chlorite Schist**
Pale grey, well-foliated, muscovite-rich.

QMCGS — **Quartz-Muscovite-Chlorite-Garnet Schist**
Pale green, fine grained, quartz-rich, well-foliated with up to 5% ragged reddish garnet porphyroblasts.

QMPS — **Quartz-Muscovite-Plagioclase Schist**
Pale green, fine grained, quartz and/or plagioclase with muscovite.

QS/QTZT — **Quartz-Schist/Quartzite**
Pale green, siliceous, muscovite-rich layers.

ARG — **Argillite**
Dark grey to black, weak to moderately graphitic, variably calcareous; often contorted with local brecciation. Typically 5% py + po.

QGS — **Quartz-Graphite Schist**
Black, strongly graphitic, quartz segregations, brecciated. Typically >5% py + po.

FELS — **Felsite**
Pale grey, mottled, very fine-grained and siliceous-looking with variable muscovite and biotite flecks. Weakly to moderately calcareous, occasional marble and calc-silicate layers. Typically 5-10% po.

MARB — **Marble**
Pale grey, well-layered, fine to medium grained calcite.

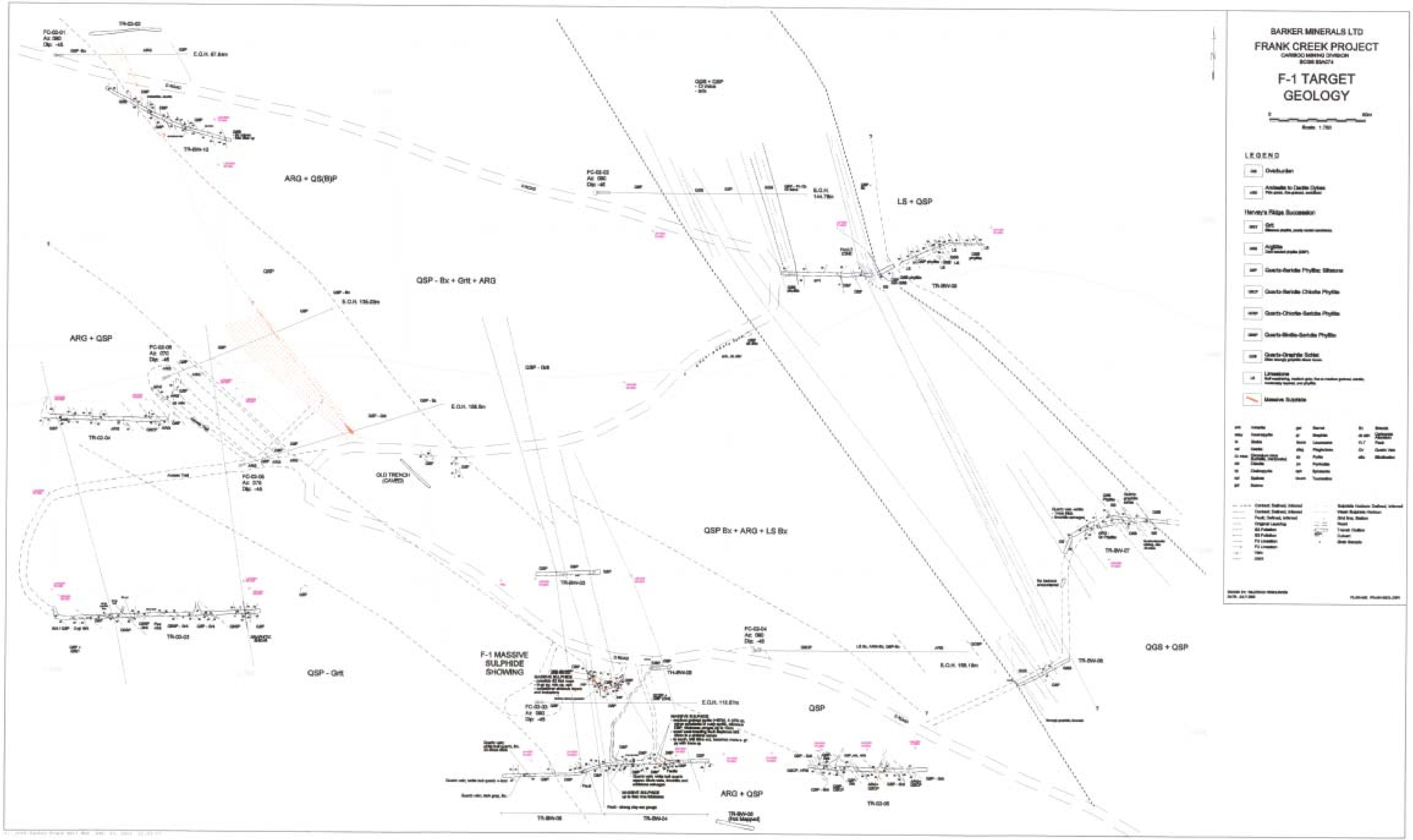
CS — **Calc-Schist**
Pale-green, moderately layered, siliceous and micaceous sections, locally strongly calcareous.

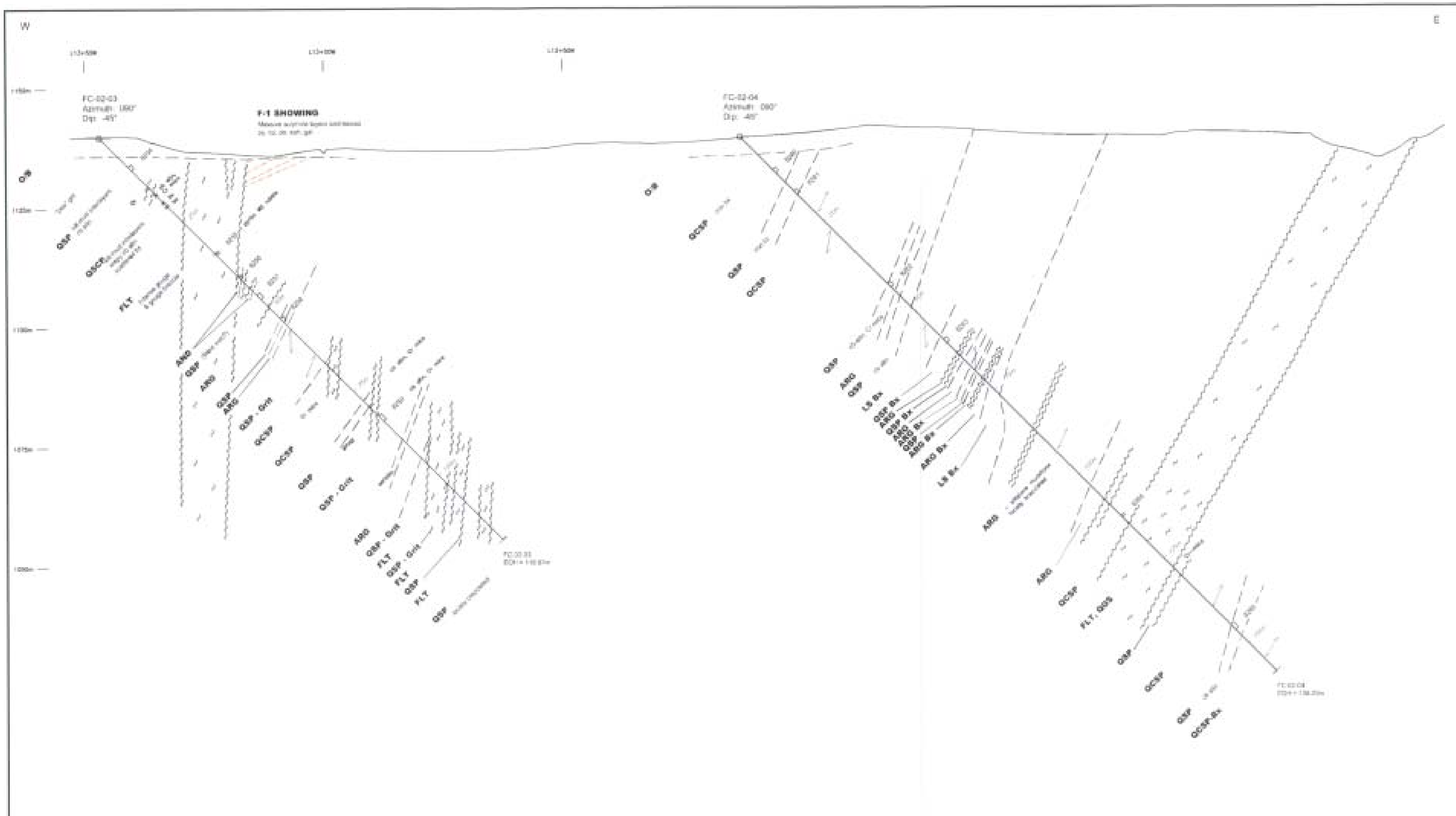
ank	Ankerite	gar	Garnet	flt	Fault
aspy	Arsenopyrite	gr	Graphite	QV	Quartz Vein
bi	Biotite	leuco	Leucoxene	silic	Silicification
cal	Calcite	plag	Plagioclase		
chl	Chlorite	py	Pyrite		
cp	Chalcopyrite	po	Pyrrhotite		
epi	Epidote	sph	Sphalerite		
gal	Galena	tourm	Tourmaline		

Foliation: S2 (principal foliation); S3
Bedding
Fold Vergence
Lineation
Contact: Defined, Inferred
Fault: Defined, Inferred
Sulphide Horizon
Weak Sulphide Horizon

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

FILENAME: DDH-ACE-02-05.DWG


BARKER MINERALS LTD
FRANK CREEK PROJECT
F-1 TARGET
GEOLOGY
LEGEND
Overburden
Andesite to Dacite Dykes
Harvey's Ridge Succession
Grit
Argillite
Quartz-Sericite Phyllite; Siltstone
Quartz-Sericite Chlorite Phyllite
Quartz-Chlorite-Sericite Phyllite
Quartz-Biotite-Sericite Phyllite
Quartz-Graphite Schist
Limestone
Massive Sulphide
TR-03-02
FC-03-01
Az: 060
Dip: -45
E.O.H. 87.8m
TR-BW-10
ARG + QS(B)P
FC-03-02
Az: 060
Dip: -45
E.O.H. 144.78m
LS + QSP
TR-BW-09
QSP - Bx + Grit + ARG
E.O.H. 135.03m
ARG + QSP
FC-03-05
Az: 070
Dip: -45
QSP - Grit
E.O.H. 156.8m
TR-03-04
FC-03-06
Az: 070
Dip: -45
OLD TRENCH
(CAVED)
QSP Bx + ARG + LS Bx
TR-BW-07
TR-BW-03
TR-03-03
QSP - Grit
F-1 MASSIVE
SULPHIDE
SHOWING
FC-03-04
Az: 060
Dip: -45
E.O.H. 155.13m
TR-BW-05
QGS + QSP
E.O.H. 115.67m
FC-03-03
Az: 060
Dip: -45
QSP
ARG + QSP
TR-03-05
TR-BW-06
TR-BW-04
TR-BW-05
(Not Mapped)



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[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]
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Tag #	Cu (ppm)	Pb (ppm)	Zn (ppm)	Ag (ppm)	As (ppm)	Sb (ppm)	Se (ppm)	Hg (ppb)	Te (ppm)	Au (ppb)	Length (m)
[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]	[illegible]
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BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

FRANK CREEK FC-02-03 & FC-02-04 SECTION

VIEW TO NORTH

SCALE 1:500

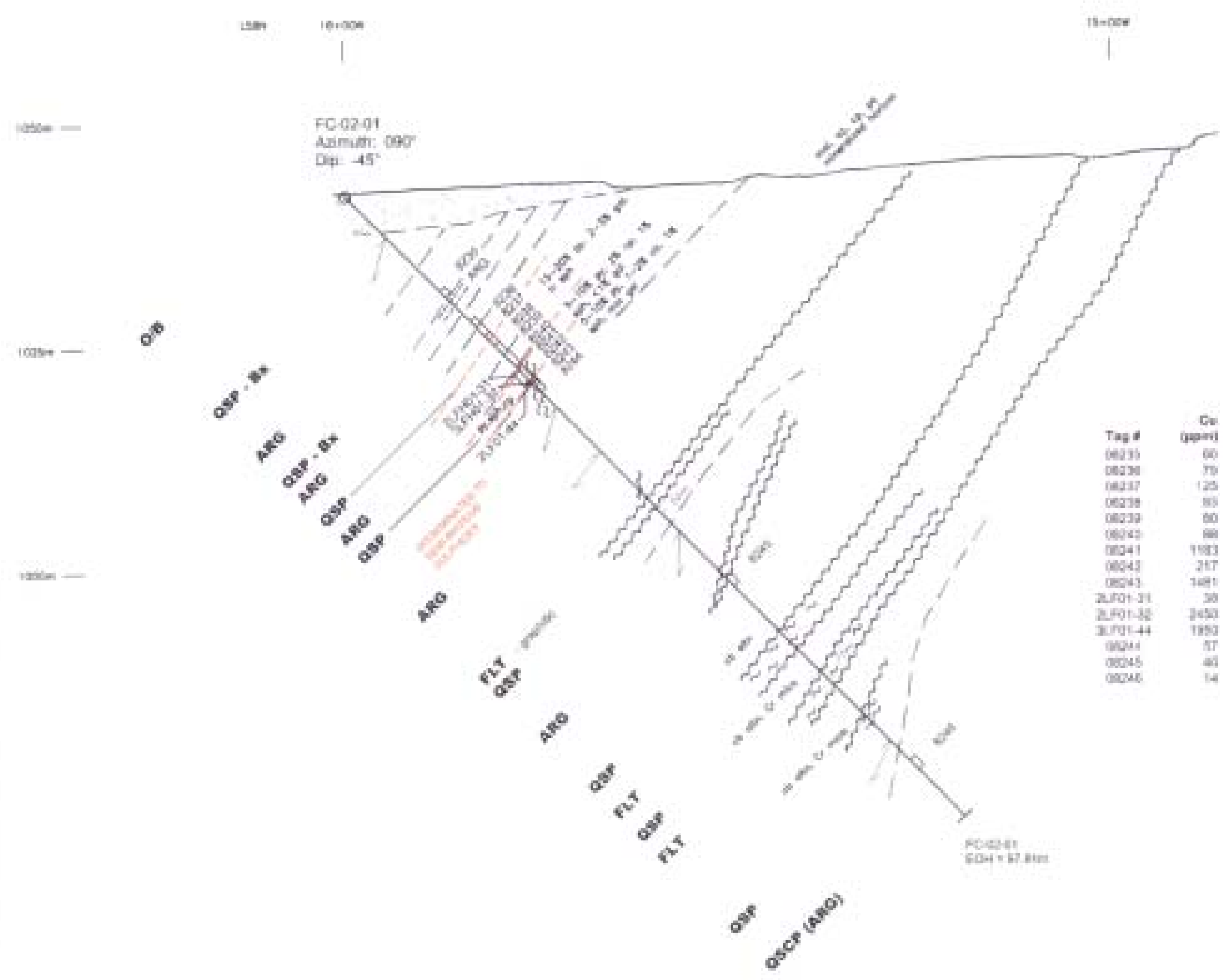
LEGEND

- OB — Overburden
- AND — Andesite to Dacite Dykes

Harvey's Ridge Succession

- Grit
- Argillite
- QSP — Quartz-Sericite Phyllite; Siltstone
- QSCP — Quartz-Sericite-Chlorite Phyllite
- QCSP — Quartz-Chlorite-Sericite Phyllite
- QBSP — Quartz-Biotite-Sericite Phyllite
- QGS — Quartz-Graphite Schist
- LS — Limestone
- Massive Sulphide

SW NE



Tag #	Cu (ppm)	Pb (ppm)	Zn (ppm)	Ag (ppm)	As (ppm)	Se (ppm)	Sn (ppm)	Hg (ppb)	Te (ppm)	Au (ppb)	Length (m)
08235	60	118	109	0.2	33	2.5	0.3	10	0.03	<5	1.00
08236	79	162	234	0.6	80	1.1	5.0	26	0.05	<5	1.00
08237	125	398	994	1.2	65	0.9	1.4	70	0.02	<5	1.40
08238	93	1374	1725	2.5	60	1.2	2.2	116	0.05	<5	1.50
08239	60	946	1065	1.0	50	1.0	1.0	80	0.03	<5	1.40
08240	68	618	2031	1.8	75	1.4	2.8	129	0.03	8	2.00
08241	1183	3226	2.30%	18.0	130	11.7	10.8	1702	0.25	13	0.40
08242	217	208	2061	1.6	50	1.3	2.3	163	0.04	5	1.80
08243	1491	908	3.40%	12.4	115	9.2	15.4	2820	0.24	20	0.40
2LF01-31	38	130	479	0.5	60	1.0	22.0	30	<0.05	<2	1.90
2LF01-32	2450	1020	3.42%	11.5	84	11.0	40.4	2350	0.25	14	0.14
3LF01-44	1950	888	3.07%	11.5	183	12.0	56.2	3040	0.30	22	0.40
08244	57	60	622	0.8	65	0.6	9.0	36	0.02	5	1.30
08245	40	932	1379	0.6	5	0.6	0.9	35	<0.02	5	1.40
08246	14	6	45	0.2	<5	0.3	0.2	<5	<0.02	<5	1.40

BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

FRANK CREEK
FC-02-01 SECTION

VIEW TO NORTH

0 5 10 15 20 25m

SCALE: 1:500

LEGEND

- O/B — Overburden
- AND — Andesite to Dacite Dykes

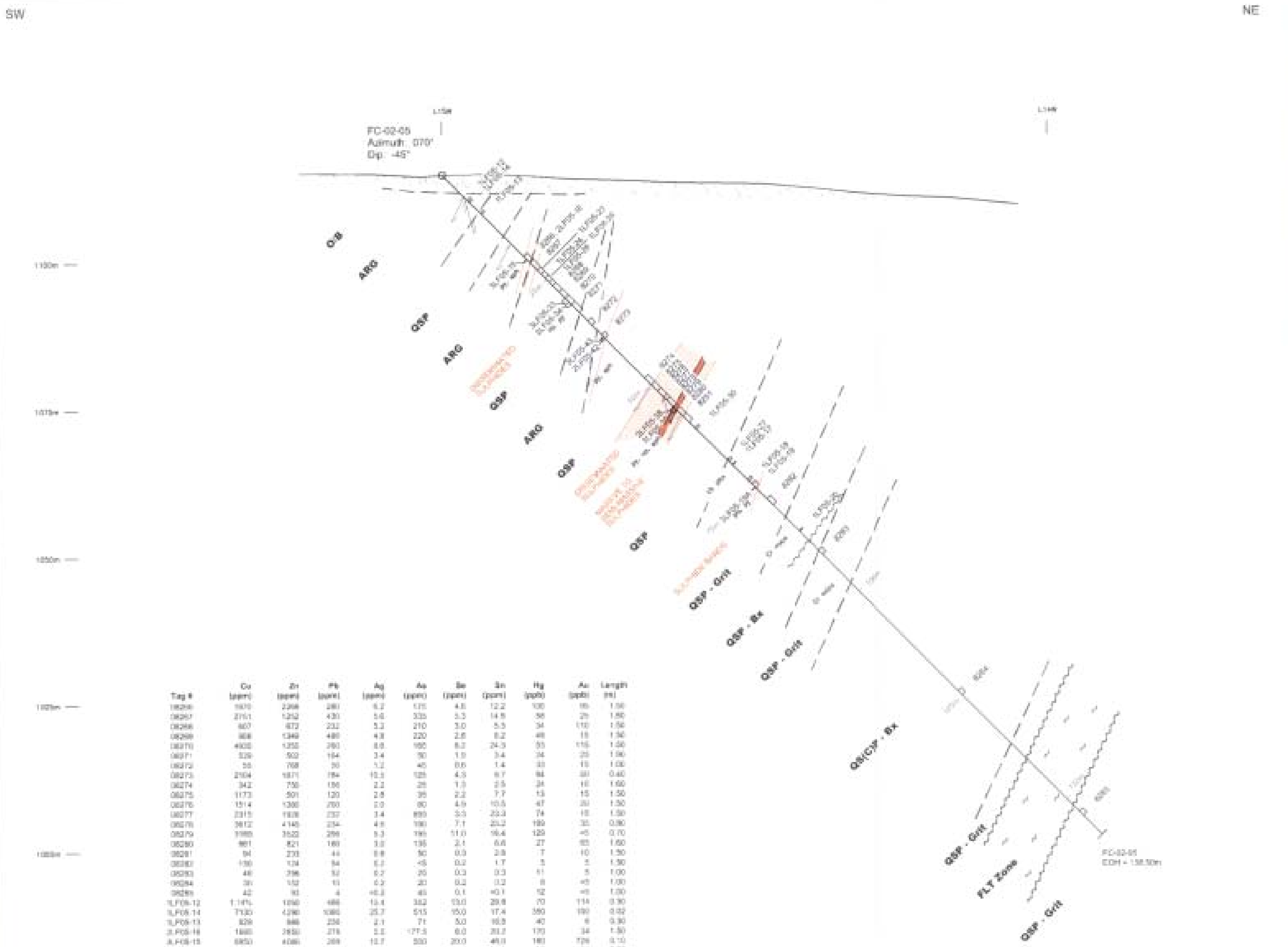
Harvey's Ridge Succession

- GRIT — Grit
- ARG — Argillite
- QSP — Quartz-Sericite Phyllite; Siltstone
- QSCP — Quartz-Sericite Chlorite Phyllite
- QCSP — Quartz-Chlorite-Sericite Phyllite
- QBSP — Quartz-Biotite-Sericite Phyllite
- QGS — Quartz-Graphite Schist
- LS — Limestone
- Massive Sulphide

ank	Ankerite	gar	Garnet	Bx	Breccia
aspy	Arsenopyrite	gr	Graphite	cb altn	Carbonate Alteration
bi	Biotite	leuco	Leucoxene	FLT	Fault
cal	Calcite	plag	Plagioclase	QV	Quartz Vein
Cr mica	Chromium Mica (fuchsite, mariposite)	py	Pyrite	silic	Silicification
chl	Chlorite	po	Pyrrhotite		
cp	Chalcopyrite	sph	Sphalerite		
epi	Epidote	tourm	Tourmaline		
gal	Galena				

- Foliation: S2 (principal foliation); S3
- Bedding
- Vein
- Fold Vergence
- Lineation
- Contact: Defined; Inferred
- Fault: Defined; Inferred
- Sulphide Horizon
- Weak Sulphide Horizon

DRAWN BY: WALDRICK RESOURCES
DATE: JULY 2002



Tag #	Cu (ppm)	Zn (ppm)	Pb (ppm)	Ag (ppm)	As (ppm)	Sb (ppm)	Sn (ppm)	Hg (ppb)	Au (ppb)	Length (m)
08266	1670	2266	280	6.2	175	4.6	12.2	100	16	1.50
08267	2751	1252	430	5.6	335	5.3	14.8	58	25	1.50
08268	607	672	232	5.2	210	3.0	5.5	34	110	1.50
08269	868	1349	480	4.8	220	2.8	8.2	48	15	1.50
08270	4600	1250	280	8.8	160	8.2	24.3	55	115	1.00
08271	329	502	164	3.4	50	1.0	3.4	34	25	1.00
08272	55	768	30	1.2	45	0.6	1.4	33	15	1.00
08273	2104	1671	784	15.3	125	4.3	9.7	84	20	0.40
08274	342	750	156	2.2	25	1.3	2.5	24	10	1.60
08275	1173	501	120	2.8	35	2.2	7.7	13	15	1.50
08276	1514	1366	260	2.0	80	4.5	10.5	47	20	1.50
08277	2315	1926	237	3.4	850	3.5	23.3	74	15	1.50
08278	3612	4146	234	4.6	190	7.1	23.2	189	30	0.90
08279	1165	3522	266	5.3	195	11.0	16.4	129	<5	0.70
08280	861	821	160	3.0	135	2.1	6.6	27	65	1.60
08281	94	233	44	0.8	50	0.3	2.8	7	10	1.50
08282	150	124	54	0.2	<5	0.2	1.7	5	5	1.50
08283	48	396	52	0.2	20	0.3	0.3	51	5	1.00
08284	30	152	13	0.2	20	0.2	0.2	8	<5	1.00
08285	42	93	4	<0.2	45	0.1	<0.1	12	<5	1.00
1LF05-12	1.14%	1050	486	15.4	352	13.0	29.8	70	114	0.30
1LF05-14	7130	4290	1080	25.7	515	15.0	17.4	350	150	0.02
1LF05-13	829	866	236	2.1	71	5.0	10.5	40	6	0.30
2LF05-16	1660	2650	278	2.2	177.5	6.0	20.2	170	34	1.50
3LF05-15	6850	4086	269	12.7	550	20.0	46.0	160	726	0.10
1LF05-23	2560	1123	377	5.8	54	4.0	20.8	50	66	0.90
1LF05-24	705	1220	462	4.5	123	3.0	16.8	80	64	1.00
1LF05-25	877	1250	486	3.4	73	2.0	21.2	50	38	1.00
1LF05-26	878	512	151	2.8	35	1.0	18.4	20	30	0.80
3LF05-33	7370	2380	575	14.6	252	12.0	38.8	120	174	0.80
2LF05-34	815	254	94	2.9	46	3.0	11.4	10	10	0.50
3LF05-43	3870	2580	1150	24.6	110	9.0	34.4	100	38	0.20
2LF05-42	1725	748	515	8.0	254	6.0	19.2	30	64	0.30
2LF05-38	866	2590	133	1.5	76	6.0	33.8	90	8	0.20
3LF05-39	4850	4030	368	6.4	208	13.0	29.8	140	84	0.50
1LF05-36	109	178	128	1.5	10	1.0	7.4	10	6	0.40
1LF05-37	295	386	147	1.8	11	1.0	45	10	6	0.40
1LF05-17	351	384	45	0.9	23	1.0	19.2	10	4	0.15
1LF05-18	582	232	108	1.9	29	1.0	17.2	10	10	0.35
1LF05-19	122	94	98	0.3	37	<1	27.8	<10	<2	1.00
3LF05-19A	564	484	500	4.5	17	5.0	23.4	30	15	0.58
1LF05-20	47	82	19	0.2	463	<1	3.8	10	<2	0.20

BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

FRANK CREEK FC-02-05 SECTION

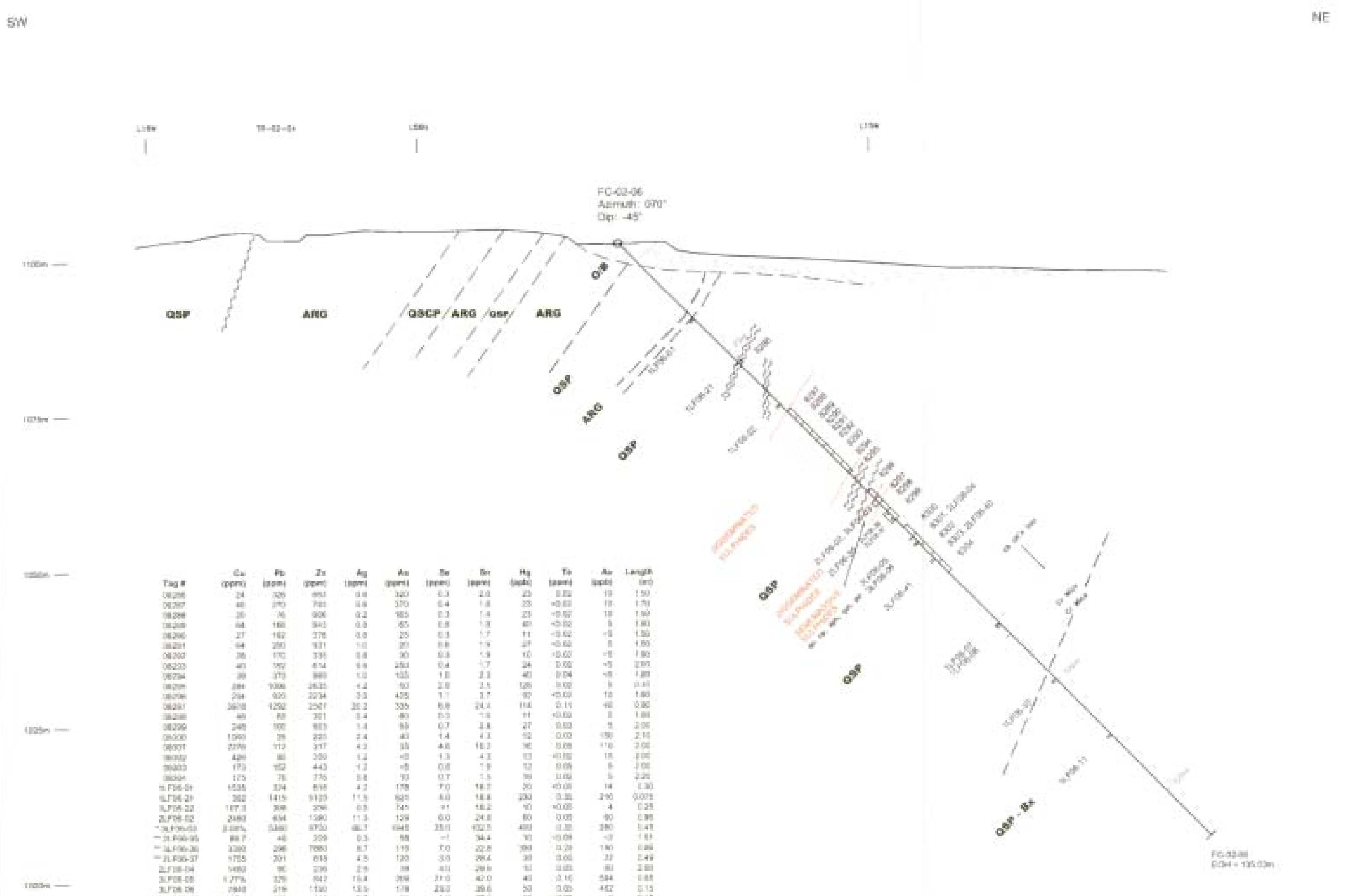
VIEW TO 345°

SCALE 1:500

LEGEND

- O/B — Overburden
- AND — Andesite to Dacite Dykes
- Harvey's Ridge Succession
- GRIT — Grit
- ARG — Argillite
- QSP — Quartz-Sericite Phyllite; Siltstone
- QSCP — Quartz-Sericite Chlorite Phyllite
- QCSP — Quartz-Chlorite-Sericite Phyllite
- QBSP — Quartz-Biotite-Sericite Phyllite
- QGS — Quartz-Graphite Schist
- LS — Limestone
- Massive Sulphide

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002



Tag #	Cu (ppm)	Pb (ppm)	Zn (ppm)	Ag (ppm)	As (ppm)	Sb (ppm)	Bi (ppm)	Hg (ppb)	Te (ppm)	Au (ppb)	Length (m)
08286	24	326	993	0.6	320	0.3	2.0	23	0.02	10	1.50
08287	48	270	792	0.6	370	0.4	1.8	23	<0.02	10	1.70
08288	26	76	906	0.2	165	0.3	1.4	23	<0.02	10	1.50
08289	64	166	943	0.5	65	0.8	1.8	40	<0.02	5	1.80
08290	27	162	378	0.5	25	0.3	1.7	11	<0.02	<5	1.50
08291	64	290	931	1.0	20	0.6	1.9	27	<0.02	5	1.80
08292	26	170	339	0.8	30	0.3	1.9	10	<0.02	<5	1.80
08293	40	192	614	0.6	250	0.4	1.7	24	0.02	<5	2.01
08294	39	273	980	1.0	105	1.0	2.3	40	0.04	<5	1.89
08295	284	1006	2635	4.2	50	2.0	3.5	126	0.02	5	0.51
08296	294	920	2234	3.3	425	1.1	3.7	92	<0.02	10	1.80
08297	2678	1292	2567	20.2	335	6.8	24.4	114	0.11	40	0.90
08298	46	69	301	0.4	80	0.3	1.6	11	<0.02	5	1.80
08299	248	100	603	1.4	55	0.7	3.8	27	0.03	5	2.00
08300	1090	39	220	2.4	40	1.4	4.3	12	0.03	150	2.10
08301	2276	113	317	4.3	35	4.8	18.2	16	0.05	110	2.00
08302	426	80	350	1.2	<5	1.3	4.3	13	<0.02	15	2.00
08303	173	102	443	1.2	<5	0.6	7.8	12	0.05	5	2.00
08304	175	76	776	0.8	10	0.7	1.5	39	0.02	5	2.25
1LF06-01	1535	324	616	4.2	178	7.0	18.2	20	<0.05	14	0.30
1LF06-21	382	1415	5120	11.5	621	4.0	18.8	230	0.35	210	0.075
1LF06-22	187.3	308	296	0.5	741	<1	18.2	10	<0.05	4	0.25
2LF06-02	2480	454	1380	11.3	129	6.0	24.8	80	0.05	80	0.95
* 2LF06-03	2.08%	5860	8730	66.7	1845	35.0	102.8	400	0.30	280	0.45
** 2LF06-35	88.7	48	229	0.3	58	<1	34.4	10	<0.05	<2	1.61
** 2LF06-36	3390	298	7880	8.7	115	7.0	22.8	390	0.20	180	0.88
** 2LF06-37	1755	201	618	4.5	120	3.0	39.4	30	0.05	22	0.49
2LF06-04	1480	96	239	2.5	39	4.0	28.6	10	0.05	80	2.83
3LF06-05	1.77%	379	847	16.4	269	21.0	42.0	40	0.10	584	0.85
3LF06-06	2848	219	1150	13.5	178	23.0	39.6	50	0.05	452	0.15
*** 2LF06-40	137.4	108	410	0.7	21	1.0	27.0	10	<0.05	2	0.15
3LF06-41	512	321	465	3.2	24	4.0	25.6	10	0.25	8	0.45
1LF06-07	1560	61	380	2.3	8	3.0	33.4	10	0.10	132	0.20
1LF06-08	991	114.5	475	2.2	48	6.0	49.0	10	0.05	36	0.33
1LF06-10	27	28.5	40	0.1	3	<1	2.2	<10	<0.05	<2	0.30
1LF06-11	22.7	14.5	50	0.1	8	1.0	8.8	<10	<0.05	8	0.30

* [illegible] of remobilization within 0.96m interval
** Samples making [illegible]
*** Sample is comprised of 0.8m & 0.7m remobilized intercept within 71.0m - 73.3m

BARKER MINERALS LTD
FRANK CREEK PROJECT
CARIBOO MINING DIVISION
BCGS 93A074

FRANK CREEK FC-02-06 SECTION
VIEW TO 340°

SCALE: 1:500

LEGEND

- O/B — Overburden
- AND — Andesite to Dacite Dykes
- Harvey's Ridge Succession
- GRIT — Grit
- ARG — Argillite
- QSP — Quartz-Sericite Phyllite; Siltstone
- QSCP — Quartz-Sericite Chlorite Phyllite
- QCSP — Quartz-Chlorite-Sericite Phyllite
- QBSP — Quartz-Biotite-Sericite Phyllite
- QGS — Quartz-Graphite Schist
- LS — Limestone
- Massive Sulphide

ank	Ankerite	gar	Garnet	Bx	Breccia
aspy	Arsenopyrite	gr	Graphite	cb alt	Carbonate Alteration
bi	Biotite	leuco	Leucoxene	FLT	Fault
cal	Calcite	plag	Plagioclase	QV	Quartz Vein
Cr mica	Chromium Mica (fuchsite, mariposite)	py	Pyrite	sil	Silicification
chl	Chlorite	po	Pyrrhotite		
cp	Chalcopyrite	sph	Sphalerite		
ep	Epidote	tourm	Tourmaline		
gal	Galena				

- Foliation: S2 (principal foliation) S3
- Bedding
- Vein
- Fold Vergence
- Lineation
- Contact: Defined, Inferred
- Fault: Defined, Inferred
- Sulphide Horizon
- Weak Sulphide Horizon

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DATE: JULY 2002

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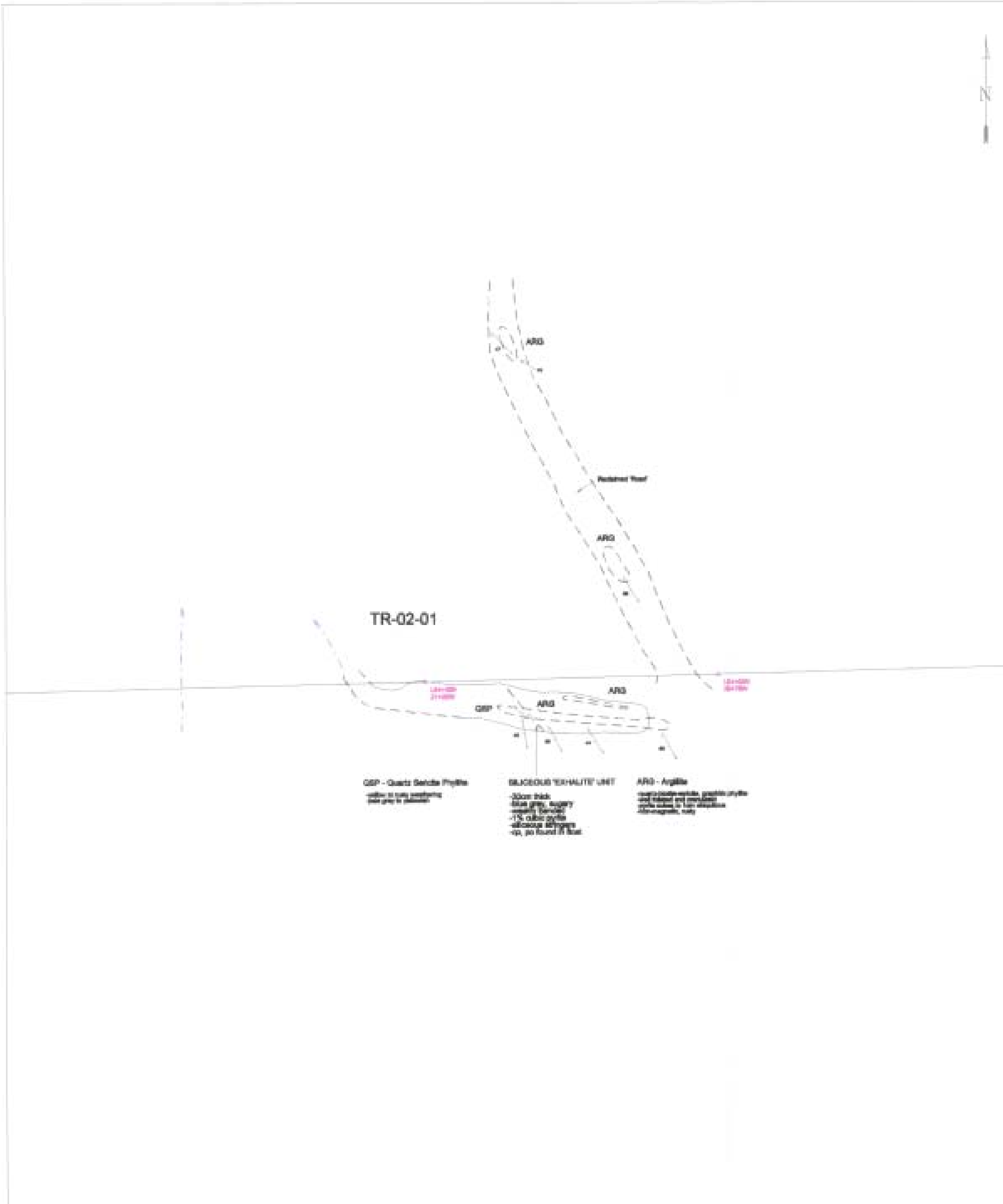

BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

TRENCH MAP
TR-02-01

0 10m

Scale: 1:250

LEGEND

O/B	Overburden
AND	Andesite to Dacite Dykes Pale green, fine-grained, sericitized

Harvey's Ridge Succession

GRIT	Grit Siliceous phyllite, poorly sorted sandstone.
ARG	Argillite Dark banded phyllite (DSP).
QSP	Quartz-Sericite Phyllite; Siltstone
QSCP	Quartz-Sericite Chlorite Phyllite
QCSP	Quartz-Chlorite-Sericite Phyllite
QBSP	Quartz-Biotite-Sericite Phyllite
QGS	Quartz-Graphite Schist Often strongly graphitic shear zones.
LS	Limestone Buff weathering, medium grey, fine to medium grained, calcitic, moderately layered, and phyllitic.
	Massive Sulphide

ank	Ankerite	gar	Garnet	Bx	Breccia
aspy	Arsenopyrite	gr	Graphite	cb altn	Carbonate Alteration
bi	Biotite	leuc	Leucoxene	FLT	Fault
cal	Calcite	plag	Plagioclase	QV	Quartz Vein
Cr mica	Chromium mica (fuchsite, mariposite)	py	Pyrite	silic	Silicification
chl	Chlorite	po	Pyrrhotite		
cp	Chalcopyrite	sph	Sphalerite		
epi	Epidote	tourn	Tourmaline		
gal	Galena				

Contact; Defined, Inferred
Fault; Defined, Inferred
Original Layering
S2 Foliation
S3 Foliation
F3 Lineation
F2 Lineation
Vein
Joint

Sulphide Horizon
Weak Sulphide Horizon
Grid line; Station
Road
Trench Outline
Culvert
Grab Sample

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

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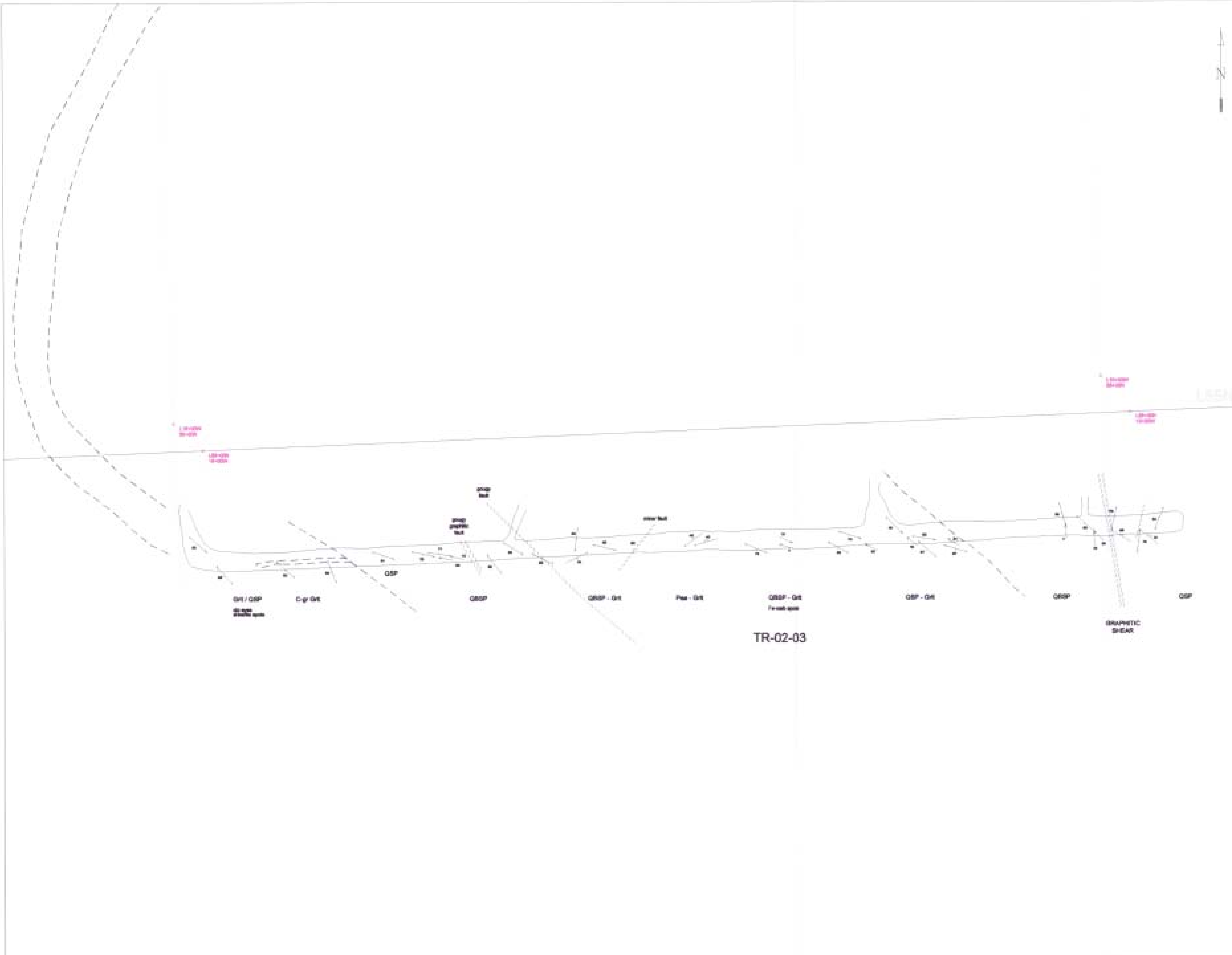

BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

TRENCH MAP
TR-02-03

0 10m
Scale: 1:250

LEGEND

O/B	Overburden
AND	Andesite to Dacite Dykes Pale green, fine grained, sericitized

Harvey's Ridge Succession

GRIT	Grit Siliceous phyllite, poorly sorted sandstone.
ARG	Argillite Dark banded phyllite (QSP).
QSP	Quartz-Sericite Phyllite; Siltstone
QSCP	Quartz-Sericite Chlorite Phyllite
QCSP	Quartz-Chlorite-Sericite Phyllite
QBSP	Quartz-Biotite-Sericite Phyllite
QGS	Quartz-Graphite Schist Often strongly graphitic shear zones.
LS	Limestone Buff weathering, medium grey, fine to medium grained, calcitic, moderately layered, and phyllitic.
	Massive Sulphide

ank	Ankerite	gar	Garnet	Bx	Breccia
aspy	Arsenopyrite	gr	Graphite	cb.altn	Carbonate Alteration
bi	Biotite	leuco	Leucoxene	FLT	Fault
cal	Calcite	plag	Plagioclase	QV	Quartz Vein
Cr mica	Chromium mica (fuchsite, mariposite)	py	Pyrite	silic	Silicification
chl	Chlorite	po	Pyrrhotite		
cp	Chalcopyrite	sph	Sphalerite		
epi	Epidote	tourm	Tourmaline		
gal	Galena				

Contact; Defined, Inferred
Fault; Defined, Inferred
Original Layering
S2 Foliation
S3 Foliation
F3 Lineation
F2 Lineation
Vein
Joint

Sulphide Horizon
Weak Sulphide Horizon
Grid line; Station
Road
Trench outline
Culvert
Grab sample

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DATE: JULY 2002

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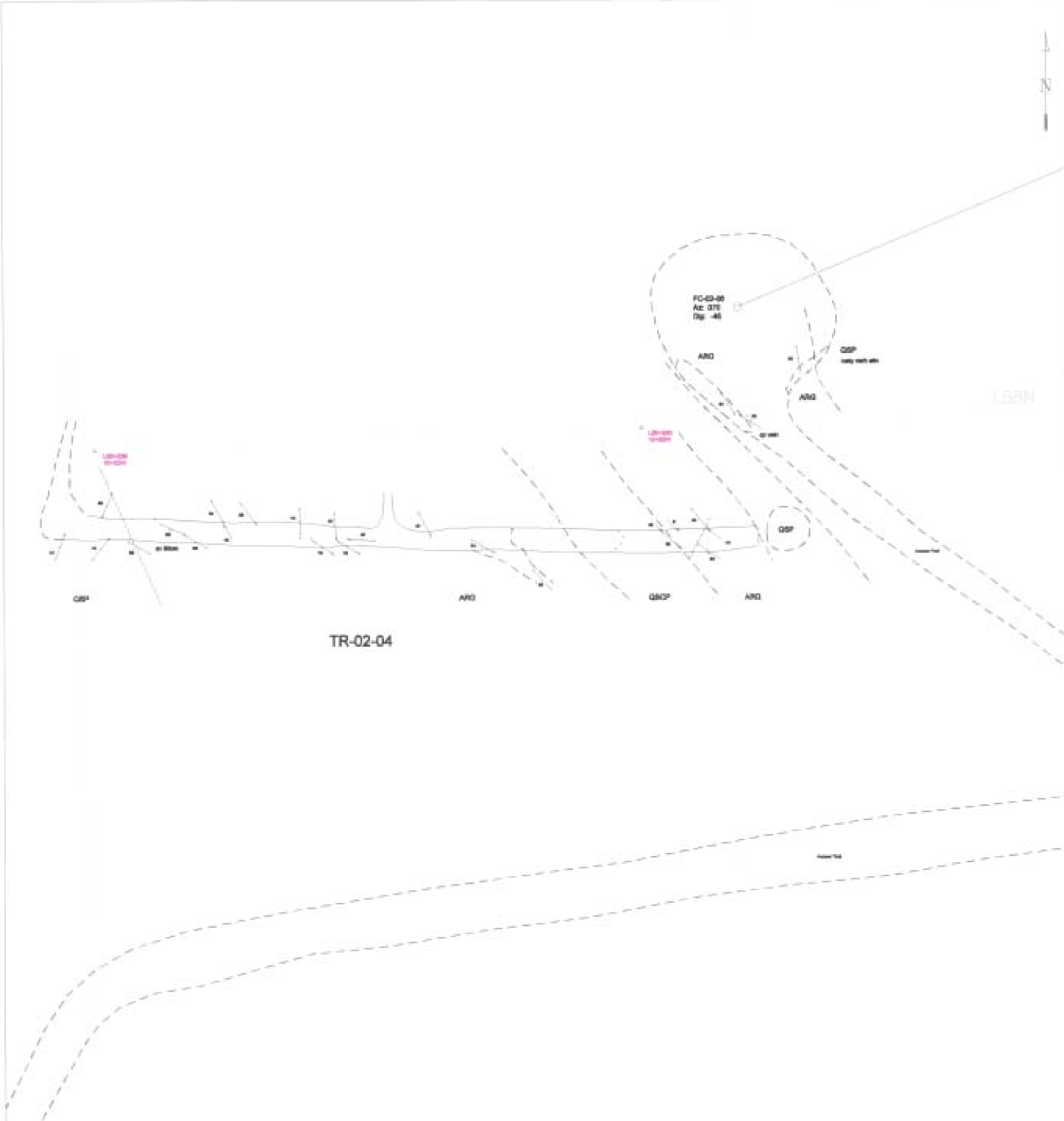

BARKER MINERALS LTD

FRANK CREEK PROJECT

CARIBOO MINING DIVISION
BCGS 93A074

TRENCH MAP
TR-02-04

0 10m
Scale: 1:250

LEGEND

- O/B — Overburden
- AND — Andesite to Dacite Dykes
  Pale green, fine-grained, sericitized

Harvey's Ridge Succession

- GRIT — Grit
  Siliceous phyllite, poorly sorted sandstone
- ARG — Argillite
  Dark banded phyllite (DBP)
- QSP — Quartz-Sericite Phyllite; Siltstone
- QSCP — Quartz-Sericite Chlorite Phyllite
- QCSP — Quartz-Chlorite-Sericite Phyllite
- QBSP — Quartz-Biotite-Sericite Phyllite
- QGS — Quartz-Graphite Schist
  Often strongly graphitic shear zones.
- LS — Limestone
  Buff weathering, medium grey, fine to medium grained, calcitic, moderately layered, and phyllitic.
- Massive Sulphide

ank	Ankerite	gar	Garnet	Bx	Breccia
aspy	Arsenopyrite	gr	Graphite	cb-altn	Carbonate Alteration
bi	Biotite	leuco	Leucoxene	FLT	Fault
cal	Calcite	plag	Plagioclase	QV	Quartz Vein
Cr mica	Chromium mica (fuchsite, mariposite)	py	Pyrite	silic	Silicification
chl	Chlorite	po	Pyrrhotite		
cp	Chalcopyrite	sph	Sphalerite		
epi	Epidote	tourm	Tourmaline		
gal	Galena				

Contact; Defined, Inferred
Fault; Defined, Inferred
Original Layering
S2 Foliation
S3 Foliation
F3 Lineation
F2 Lineation
Vein
Joint

Sulphide Horizon
Weak Sulphide Horizon
Grid line; Station
Road
Trench outline
Culvert
Grab sample

DRAWN BY: WILDROCK RESOURCES
DATE: JULY 2002

FILENAME: TR-02-04.DGN

A PRELIMINARY REPORT

ON

ELECTROMAGNETIC, GRAVITY, MAGNETIC &

INDUCED POLARIZATION SURVEYING

Ace & Frank Creek Properties
Cariboo Lake Area, B.C.
52° 45'N, 121° 15'W
N.T.S. 93 A 14

For

BARKER MINERALS LIMITED

Langley, British Columbia

By

PETER E. WALCOTT & ASSOCIATES LIMITED

Vancouver, British Columbia

SEPTEMBER 2002

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INTRODUCTION.

Between May 12th and June 8th, 2002, Peter E. Walcott & Associates Limited undertook a geophysical surveying programme over portions of two properties for Barker Minerals Limited.

The properties, known as the Ace, & Frank Creek respectively, are located in the Cariboo Mining District of British Columbia, on average 25 kilometres northeast of the settlement of Likely.

The programme consisted mainly of horizontal loop electromagnetic (E.M.) and gravity surveying on freshly hand cut picket lines on extensions of the grids from the 2001 surveys.

In addition some VLF EM traversing was carried out around the Frank Creek showing, a small mise-a- la-masse survey was undertaken over the same showing with the current electrode implanted in it, a short I.P. traverse was conducted over the projection of the strike of the mineralization on Line 5800N, and some detailed magnetic surveying was done over portions of the Ace grid.

The data are presented at a scale of 1:2500 in various forms on maps that accompany this report.

PROPERTY, LOCATION & ACCESS

The property consisting of 3949 units is located in the Cariboo Mining District of British Columbia, and stretches from the Likely to the Quesnel area.

The individual properties, known as the Ace & Frank Creek respectively, are situated between the Caribou and Quesnel lakes some 20 to 45 kilometres northeast of the settlement of Likely.

Access was obtained from Likely by means of 4-wheel drive vehicle along the logging roads that traverse the properties.

GEOLOGY.

The Ace & Frank Creek properties are located in the geological terrane known as the Barkerville Terrane, fault-bounded on the west from the Quesnel Terrane by the Eureka Thrust, and similarly on the east from the Cariboo Terrane by the Pleasant Valley Thrust.

The Barkerville Terrane is classified by the GSC as a subset of the Kootenay Terrane, deposited along the western edge of ancient North America.

Metamorphosed sedimentary and volcanic rocks - the result of bimodal volcanism in an island-arc scenario in the Late Proterozoic to Mid-Paleozoic age - underlie the property. These have been deformed by intense tectonism that has caused complex folding and overturning.

Volcanic–associated massive sulphide deposits occur in submarine volcanic rocks throughout the world in close association with at the least minor sedimentary assemblages. Thus the property areas are thought to have potential to host a polymetallic massive sulphide deposit(s)

A one metre thick massive copper-zinc-lead-silver sulphide occurrence was observed in black phyllites on the Frank Creek property. Additional pods of pyrite-chalcopyrite mineralization were also noted in the same assemblage, as well as numerous massive sulphide boulders of mostly pyrite.

A mineralized boulder field some 8 kilometres long by 1 kilometre wide trends roughly east-west through the Ace property. In it some 1000 mostly iron rich angular shaped boulders have been documented and it is referred to as the Doyle Train. Values of up to 20 g/t gold and 13% combined lead-zinc with 0.25% copper and 3 oz g/t silver were contained in sulphide bearing quartz veins and iron rich massive sulphides examples of the above. Limited trenching in some areas within the boulder field has uncovered similar mineralization in bedrock.

GEOLOGY cont'd

For more detailed information the reader is referred to the numerous publications on the area and to reports on the properties held by Barker Minerals Limited, particularly the qualifying for listing by Bruce J. Perry Ph.D. P. Geo. & John G. Payne Ph.D. and to the trenching and mapping by Christopher J. Wild P.Eng.

PREVIOUS WORK.

Previous work carried out on part or all of the properties consisted of prospecting and geological mapping, geochemical soil sampling, heliborne magnetic and electromagnetic surveying, ground magnetic and VLF EM surveying, induced polarization surveying, E-scan sounding and diamond drilling.

For further details the reader is referred to the aforementioned reports on the properties and to the reports on the 2000 survey and 2001 geophysical surveys by the writer.

PURPOSE.

The purpose of the electromagnetic surveys was to locate and/or further define E.M. conductors on the properties which could be attributable to massive sulphide mineralization whereas that of the gravity was to assist in the discrimination of graphitic and sulphide conductors, based on the premise that those conductors attributable to massive sulphides would exhibit a positive density contrast with the host and surrounding rocks, resulting in discernible gravity highs due to the excess mass.

SURVEY SPECIFICATIONS.

Electromagnetic Surveying.

The basic principle of any electromagnetic survey is that when conductors are subjected to primary alternating fields secondary magnetic fields are induced in them. Measurements of these secondary fields give indications as to the size, shape, conductivity and depth of burial of conductors. In the absence of conductors no secondary fields are obtained.

The electromagnetic survey was carried out using two systems, namely a Max-Min IIA horizontal loop system, and a Geonics VLF system.

The Max-Min IIA system was manufactured by Apex Parametrics of Metropolitan Toronto, Ontario.

Readings of the inphase and quadrature components of the secondary field were made with the coils in the coplanar mode, i.e. maximum coupled, every 25 metres along the picket lines at frequencies of 222, 444, 888 and 1776 Hz.

A coil separation of 100 metres was employed on the Frank Creek F7, whereas one of 200 metres was used on the Ace grids.

Corrections for topography were made using the % slope between each 25 metre station. In all some 9.7 kilometres of E.M. traversing were carried out with the various coil separations.

The VLF system was built by Geonics Limited of Metropolitan, Toronto, Ontario. The unit makes use of the VLF transmitting stations operating for communication with submarines for its transmittal signal – the vertical antenna currents create concentric horizontal magnetic fields – and measures the vertical components of the secondary fields created as above. Here the Seattle transmitter was used.

SURVEY SPECIFICATIONS cont'd

Magnetic Surveying

The magnetic survey was carried out using a EDA - 19 proton precession magnetometer manufactured by EDA Instruments of Metropolitan Toronto, Ontario. This instrument measures variations of the earth's magnetic field to an accuracy of +/- 0.5 nanoteslas. Corrections for diurnal variations were made by comparison with readings taken at 10-second intervals on a similar instrument, held fixed at one location.

Induced Polarization Surveying.

The induced polarization (I.P.) survey was conducted using a pulse type system, the principal components of which are manufactured by Iris Instruments of Orleans, France.

The system consists basically of three units, a receiver (Iris), transmitter (Iris) and a motor generator (Generac). The transmitter, which provides a maximum of 4.0 kw d.c. to the ground, obtains its power from a 6.5 kw single phase alternator driven by a gasoline engine. The cycling rate of the transmitter is 2 seconds "current-on" and 2 seconds "current-off" with the pulses reversing continuously in polarity. The data recorded in the field consists of careful measurements of the current (I) in amperes flowing through the current electrodes C_1 and C_2, the primary voltages (V) appearing between any two potential electrodes, P_1 through P_7, during the "current-on" part of the cycle, and the apparent chargeability, (M_a) presented as a direct readout in millivolts per volt using a 200 millisecond delay and a 1000 millisecond sample window by the receiver, a digital receiver controlled by a micro-processor – the sample window is actually the total of ten individual windows of 100 millisecond widths.

The apparent resistivity (ρ_a) in ohm metres is proportional to the ratio of the primary voltage and the measured current, the proportionality factor depending on the geometry of

SURVEY SPECIFICATIONS cont'd

the array used. The chargeability and resistivity are called apparent as they are values, which that portion of the earth sampled would have if it were homogeneous. As the earth sampled is usually inhomogeneous the calculated apparent chargeability and resistivity are functions of the actual chargeability and resistivity of the rocks.

The survey was carried out using the "pole-dipole" method of surveying. In this method the current electrode, C_1, and the potential electrodes, P_1 through P_7, are moved in unison along the survey lines at a spacing of "a" (the dipole) apart, while the second current electrode, C_2, is kept constant at "infinity". The distance, "na" between C_1 and the nearest potential electrode generally controls the depth to be explored by the particular separation, "n", traverse.

On the traverse along the portion of Line 5800 near the "D" road a 25 metre dipole was employed, and first to sixth separation measurements were obtained.

Gravity Surveying.

The gravity survey was carried out using a Lacoste G gravimeter. This instrument measures variations in the earth's gravitational field to an accuracy of plus or minus 0.05 mgals, and is thermostated to eliminate thermal drift corrections.

Values of observed gravity were obtained every 25 metres along the traverses at points located by 6" spikes driven flush with the ground and flagged. Corrections for meter drift were made by tying in to previously established base stations (2001) at intervals not exceeding three hours.

The elevations of the stations were obtained to 6 centimetre accuracy using a Sokkia total station and a prism reflector. The importance of station elevation accuracy is better understood by noting the rule of thumb correlation between observed gravity and elevation change namely 3 cms elevation difference makes for a 0.01 mgal observed gravity difference.

SURVEY SPECIFICATIONS cont'd

Station positioning was converted to UTM co-ordinates for latitude correction by tying-in to points on the grids established with an Ashtech Z Surveyor GPS System from the known point the BGACS station at Williams Lake.

In all some 8.5 kilometres of surveying were carried out using these methods.

DISCUSSION OF RESULTS

These should be studied in conjunction with the results of the 2000 and 2001 work in reports written by the writer.

Frank Creek.

The VLF EM survey conducted in the vicinity of the "massive sulphide" showing showed the latter to exhibit a weak response as can be seen on Figure 16, a plot of the inphase and quadrature profiles.

Several conductor axes, located by inphase crossovers on the various line traverses, can be seen trending northwestwards across the area surveyed, in keeping with the trend of the geology.

The mise-a-la-masse survey gave a fairly concentric equipotential pattern around the buried electrode – Figure 15, but did show some elongation about a similar northwest axis. Low potentials were obtained over conductor "B" – 2001 report – to the east of the showing.

Conductor "B" & "C" are clearly discernible on the calculated resistivity from the induced polarization traverse conducted along Line 5800N. The western edge of Conductor "B", located on the eastern extremity of the traverse, gave resistivity values of 2 to 5 ohm-metres, similar to those obtained on it on the 2001 road survey.

Conductor "C", which cuts the line at an oblique angle, is represented by resistivities of 8 to 12 ohm-metres between 1760 and 1860W.

No high chargeability values are directly associated with these resistivity lows except for some near surface effects over Conductor "C" suggesting the conductors to be more shear than graphite related.

Several zones of moderate chargeabilitycan be seen on the pseudosection over a moderately high background of 25 to 30 mV/V. Only two of these have any appreciable

DISCUSSION OF RESULTS cont'd

width and depth extent, namely one between 1525 and 1600W, and another between 1725 and 1800W, with both associated with lower resistivities.

The Max-Min survey over the two lines on the F7 grid extended the wide conductor of moderate conductivity 100 metres in either direction – Figure 5 & 6.

The gravity profile run over the central line – Line 2500W – of the grid showed no residual high i.e. no excess mass, associated with the conductor, leading the writer to believe its causative source was primarily graphitic material.

Ace.
Ace N - S

The Max-Min survey showed the grid to exhibit a wide zone of complex moderate to high conductivity on the three northernmost lines – Figure 1 & 2. In fact it appears the zone could actually be three individual conductors as shown – would need additional surveying with a smaller coil separation to confirm . They appear to be folded between Lines 1400 and 1500S, and offset by faulting between Lines 1600 and 1700S where they also start to drop off in conductivity. Similar offsets can be observed between Lines 1800 and 1900S, and Lines 1900 and 2000S – see 2000 survey report.

The gravity survey – calculated using a density of 2.65 gm/cc based on rock measurements conducted last winter – showed a small 0.25 mgal residual associated with the middle of the conductive zone on Lines 1500 and 1600S, and slightly lower residuals offset to the east on Lines 1400S, and to the west – the foot wall side – on Lines 1700 and 1800S.

The magnetic survey – Figure 12 - showed no correlation between the magnetic responses and the EM conductors. In fact, the magnetic feature appears attributable to small near surface features.

DISCUSSION OF RESULTS cont'd

Drill hole Ace-02-01 finished before the geophysics was completed, intersected no mineralization of significance but appears to have been drilled into the fault between Lines 1600 and 1700S.

Drill hole Ace-02-02, drilled under the same circumstances, to test the magnetic feature projected northwards from Line 1900S on the 2000 survey, also failed to intersect mineralization of significance. Magnetic surveying on Line 1800S failed to substantiate the aforementioned projection.

Ace N-N

The Max-Min survey located a conductor of poor to moderate conductivity trending across most of the grid but petering out to the south – Figures 3 and 4. The conductor exhibits some width, circa 10 metres, on Lines 400 and 500N, and has poorer conductive material associated with its foot wall as noted on the quadrature response.

The gravity survey – calculated using a density of 2.65 gm/cc – showed a weak residual anomaly associated with the foot wall of the conductor on Lines 400 and 500N.

The residual on Line 600 and 700 N coincident with the hanging wall of the conductor appears more related to topography as can be seen by comparison with the topographic profile.

Drill hole Ace-02-03 and 05 would have been drilled into the foot wall of the conductor with the former collared just about on it, while 04 should have intersected the tail of the conductive zone near 400N. This was borne out by the drill results where hole 03 intersected some semi massive to massive sulphides at the top of the hole, and hole 04 hit 10 metres of mineralized felsite at 43.8 metres.

DISCUSSION OF RESULTS cont'd

Again magnetic surveying showed no definitive correlation with conductivity. Higher susceptibilities were associated with the foot wall of the conductor as shown by the magnetic response circa 600W between 350N and 550N – Figure 10.

Ace N-C

Five lines of magnetic traversing were undertaken on this grid with an area of higher magnetic response observed trending across the grid – Figure 11 – to the east of the weakly conductive response – 2000 report.

SUMMARY, CONCLUSIONS & RECOMMENDATIONS.

Between May 12th and June 8th, 2002, Peter E. Walcott & Associates Limited undertook Max–Min E.M., gravity, magnetic, VLF EM and induced polarization surveying for Barker Minerals Ltd. on their Cariboo property, located in the Likely area of British Columbia.

These geophysical surveys were limited to two specific areas of the property, namely the Frank Creek and Ace North grids.

As before the Frank Creek showing showed little or no response to either two geophysical techniques – mise-a-la-masse and VLF EM – leading the writer to believe that no massive sulphide mineralization of significant strike length and/or size is present.

The limited I.P. surveying on Line 5800N on the Frank Creek grid confirmed the location of two previously located (2000 EM survey) E.M. conductors but gave little encouragement for additional drill targets.

The Max-Min survey extended the broad east-west striking F-7 conductor a further 100 metres in either direction, but the ensuing gravity traverse across the central portion failed to yield any significant residual anomaly indicative of excess mass, invoking the conclusion that the conductivity is mostly attributable to graphitic material.

The Max-Min survey extended the wide conductor located on Line 1600 S on the Ace North grid in 2001 to the north and south. The results suggested the conductor to be complex, and offset by folding and/or faulting, with better conductivity – moderate to good – on the northernmost lines.

Weak residual gravity anomalies were obtained associated with the conductor axes suggesting that mineralization could be the partial source of the conductivity, but not indicating the presence of significant VMS mineralization.

SUMMARY, CONCLUSIONS & RECOMMENDATIONS cont'd

Drilling carried out before the geophysics was done appeared to have been directed into the fault offset of the conductors, and intersected no mineralization of significance.

The Max-Min survey also extended the conductor located on Line 500N on the Ace North grid in 2001 to the north and south. Here the results indicated that the conductor exhibited poor to moderate conductivity, narrowed to the north and tailed off to the south.

Weak residual gravity anomalies were obtained associated with the foot wall of the conductor on the centre lines – 400 and 500N. These could be indicative of weak mineralization but in no way suggest the presence of VMS mineralization of any significant size.

Drilling carried out here again before the completion of geophysics collared one hole in the conductor, directed another across it with encouraging results, and cored a third into the foot wall to test the gravity residual on Line 500N obtained on the 2001 survey.

Although no excess mass of significant size has been detected on the Ace grid to date, drilling has indicated the presence of favourable rock type to host such deposits. Thus the writer would recommend the continuation of the Max-Min EM – gravity programme in conjunction with geochemistry along strike across the favourable horizon(s) in the search for VMS mineralization.

SUMMARY, CONCLUSIONS & RECOMMENDATIONS cont'd

Gravity profiling could be carried out at 200 metre line spacing over the defined E.M. conductors. Surface elevation maps might have to be prepared for the application of terrain corrections in the northerly and southerly parts of the trend.

Respectfully submitted,

PETER E. WALCOTT & ASSOCIATES LIMITED

Peter E. Walcott, P.Eng.
Geophysicist

Vancouver, B.C.
September 2002

CERTIFICATION

1. I am graduate of the University of Toronto in 1962 with a B.A.Sc. in Engineering Physics, Geophysics Option.

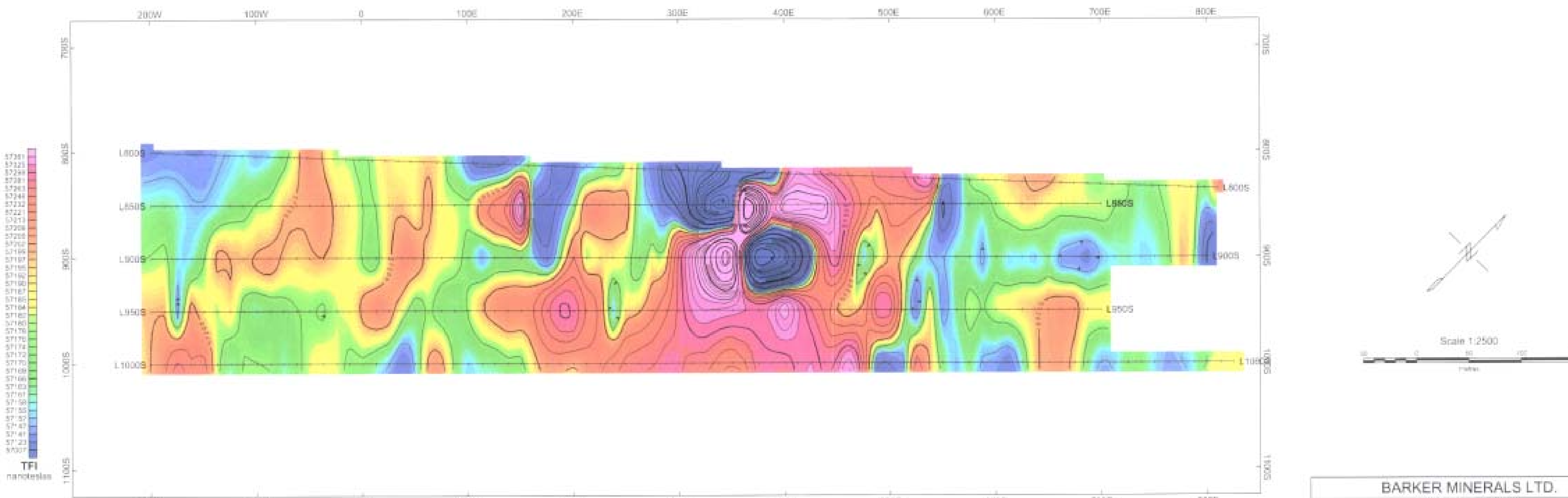
2. I have been practicing my profession for the last forty years.

3. I am a member of the Association of Professional Engineers of British Columbia and Ontario.

Peter E.Walcott, P.Eng.

Vancouver, B.C.
September, 2002

APPENDIX

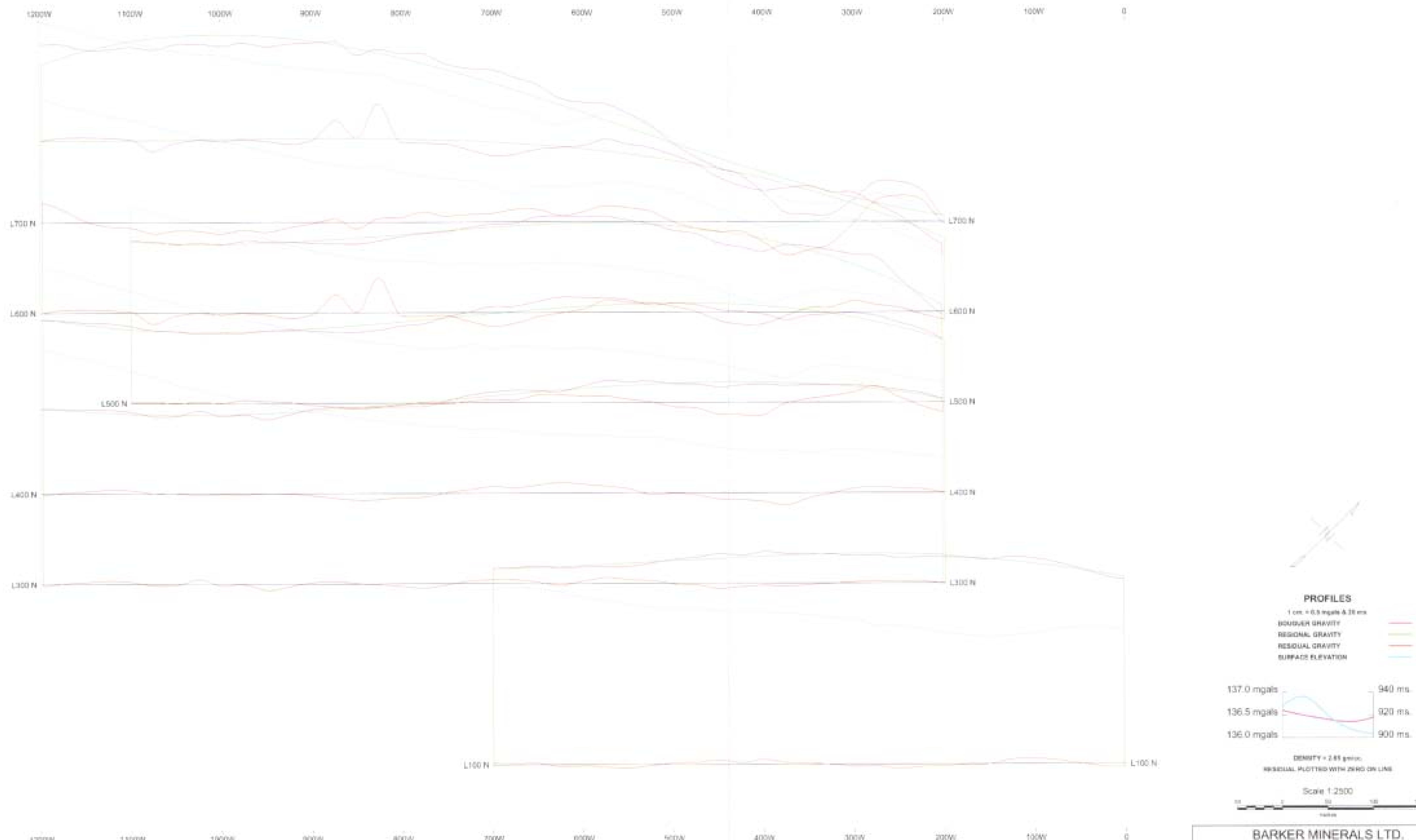


BARKER MINERALS LTD.

MAGNETIC SURVEY
CONTOURS OF TOTAL FIELD INTENSITY
IN NANOTESLAS

ACE N. C GRID
CARIBOO M.D., B.C.
FIGURE 11

PETER E. WALCOTT & ASSOCIATES LIMITED

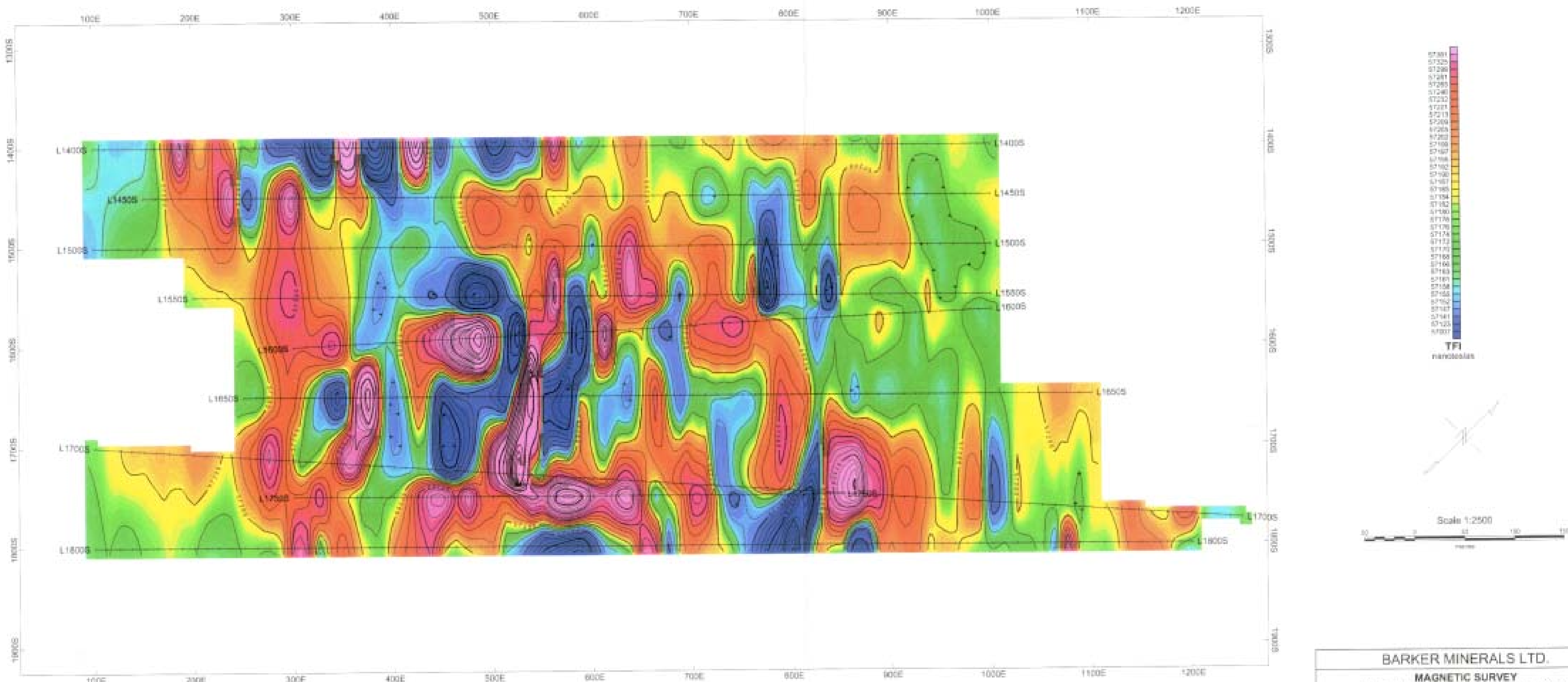


BARKER MINERALS LTD.

GRAVITY SURVEY
PROFILES OF BOUGUER GRAVITY &
SURFACE ELEVATION

ACE NORTH-N
LIKELY AREA, CARIBOU M.D., B.C.

PETER E. WALCOTT & ASSOCIATES LIMITED

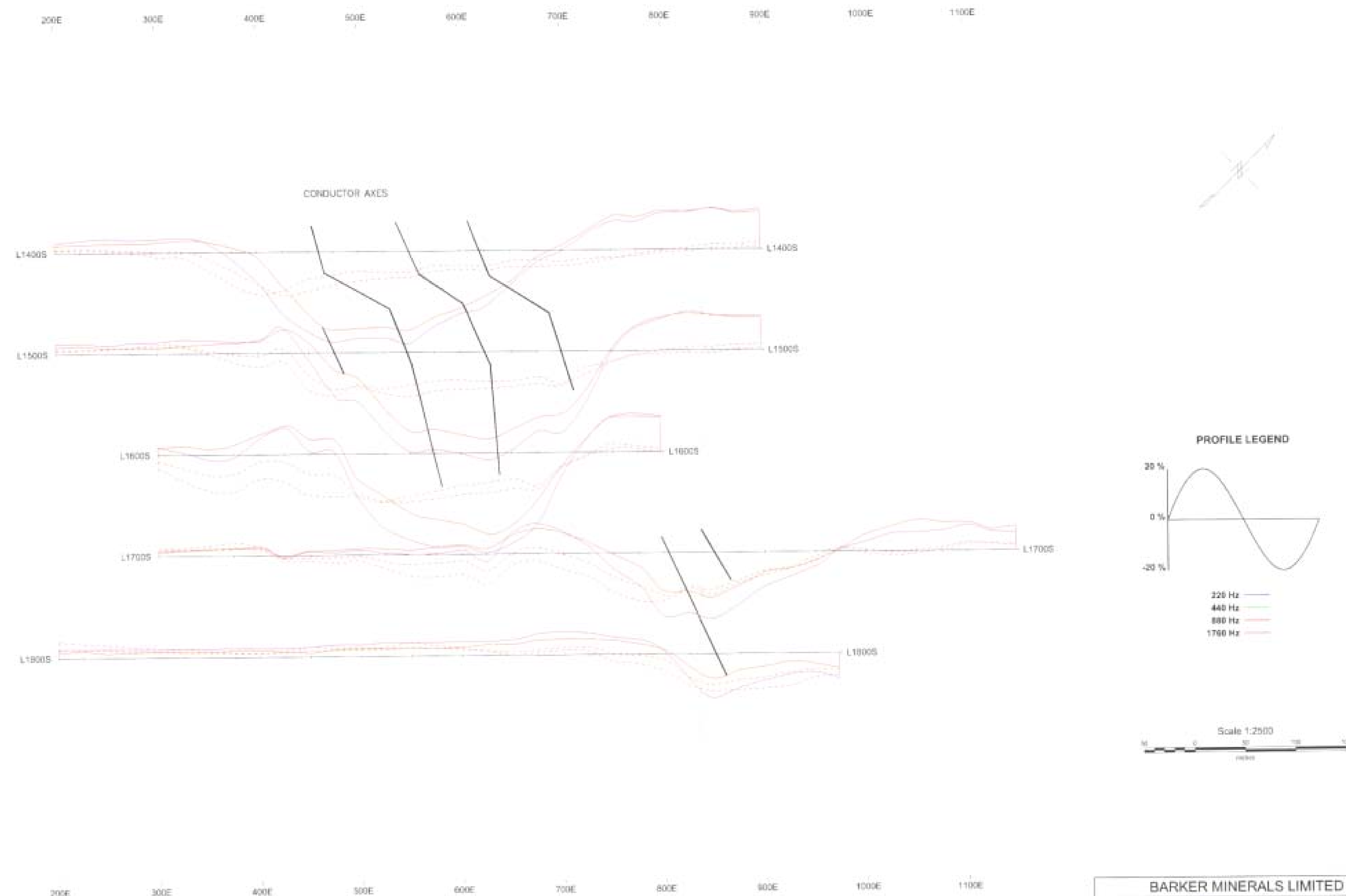


BARKER MINERALS LTD.

**MAGNETIC SURVEY
CONTOURS OF TOTAL FIELD INTENSITY
IN NANOTESLAS**

ACE N_S GRID
CARIBOO M.D., B.C.
FIGURE 12

PETER E. WALCOTT & ASSOCIATES LIMITED

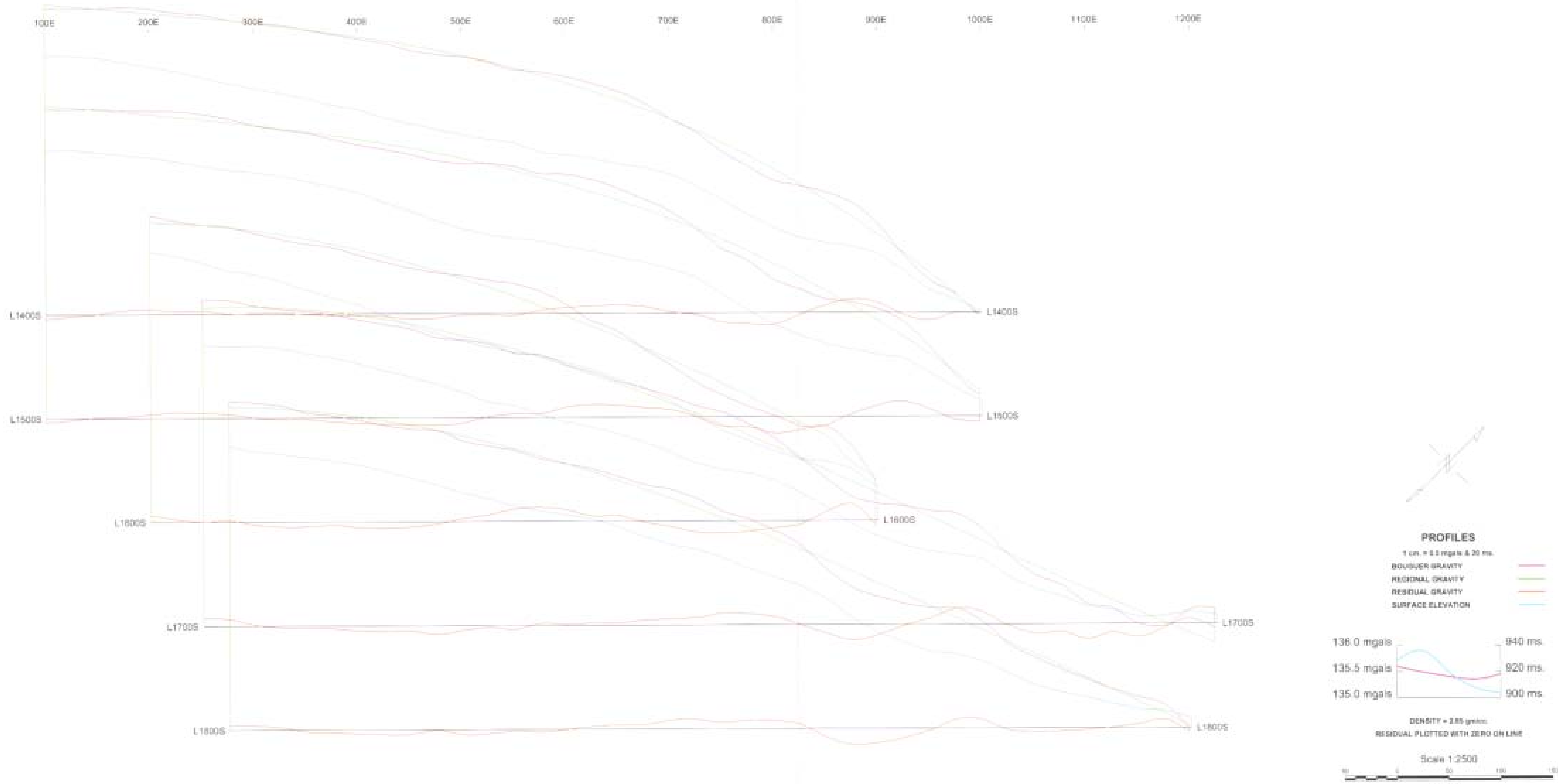


BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

ACE NORTH-5 GRID, CARIBOO M.D.
Figure: 1 Date July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED

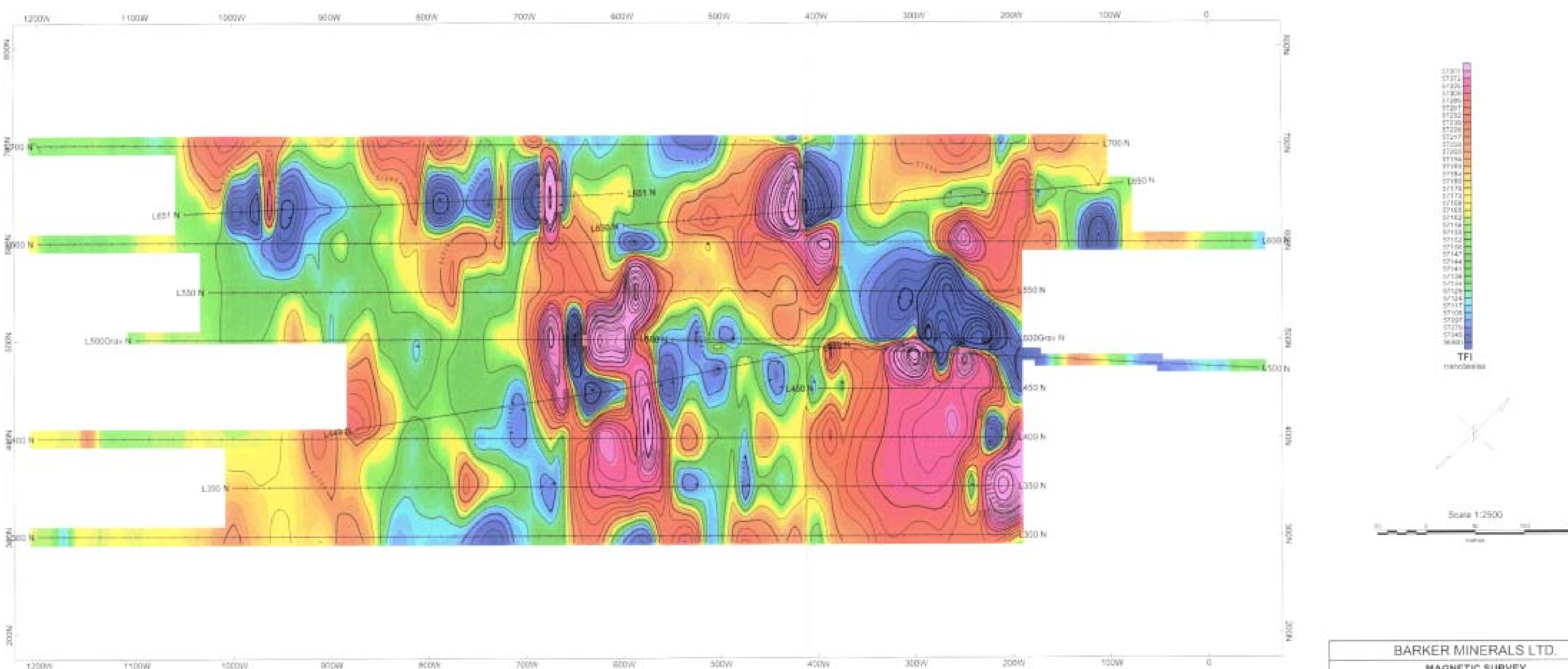


BARKER MINERALS LTD.

GRAVITY SURVEY
PROFILES OF BOUGUER GRAVITY &
SURFACE ELEVATION

ACE NORTH-S
LIKELY AREA, CARIBOU M.D., B.C.

PETER E. WALCOTT & ASSOCIATES LIMITED

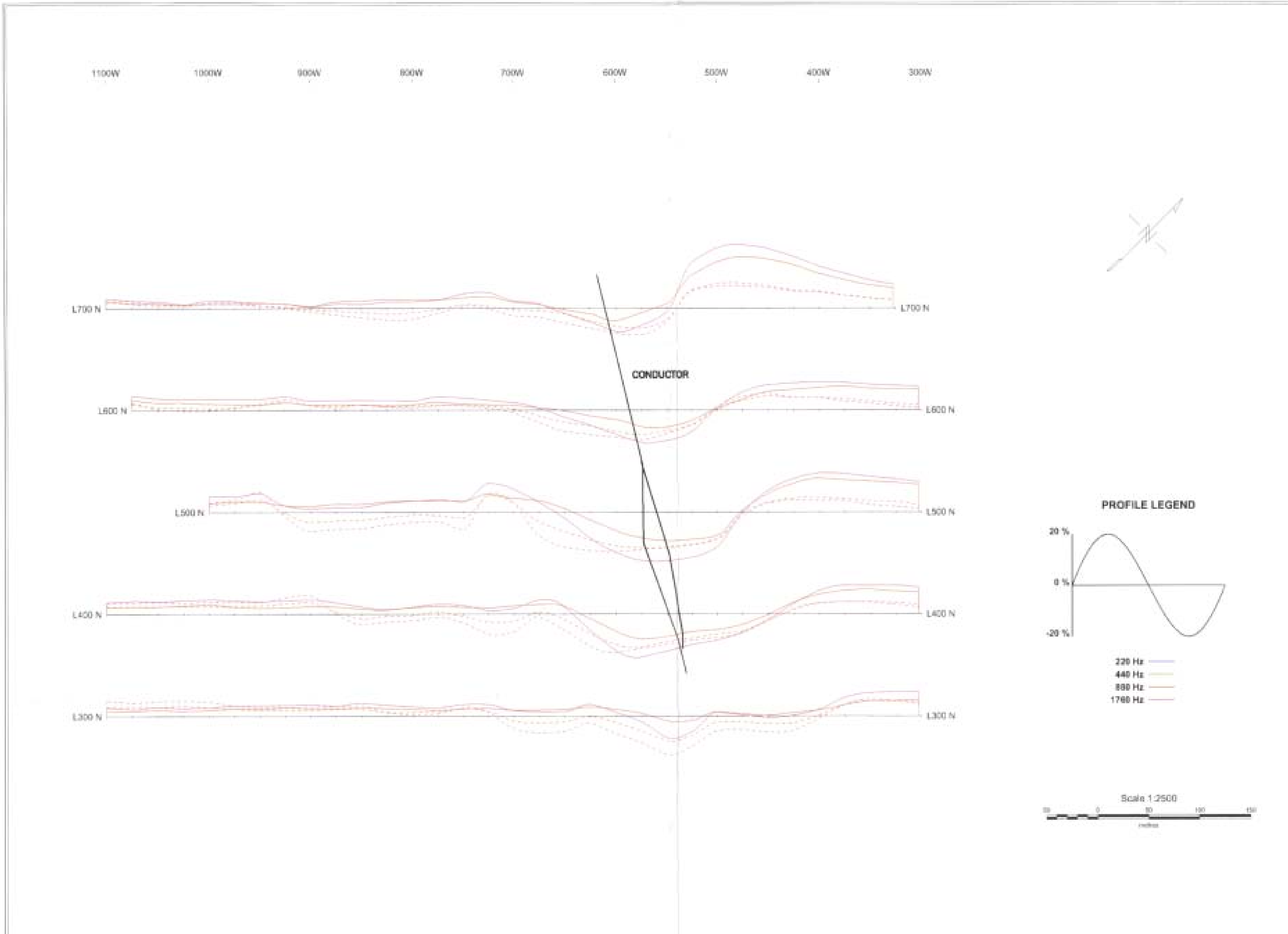


BARKER MINERALS LTD.

MAGNETIC SURVEY
CONTOURS OF TOTAL FIELD INTENSITY
IN NANOTESLAS

ACE N_N GRID
CARIBOO M.D., B.C.
FIGURE 10

PETER E. WALCOTT & ASSOCIATES LIMITED

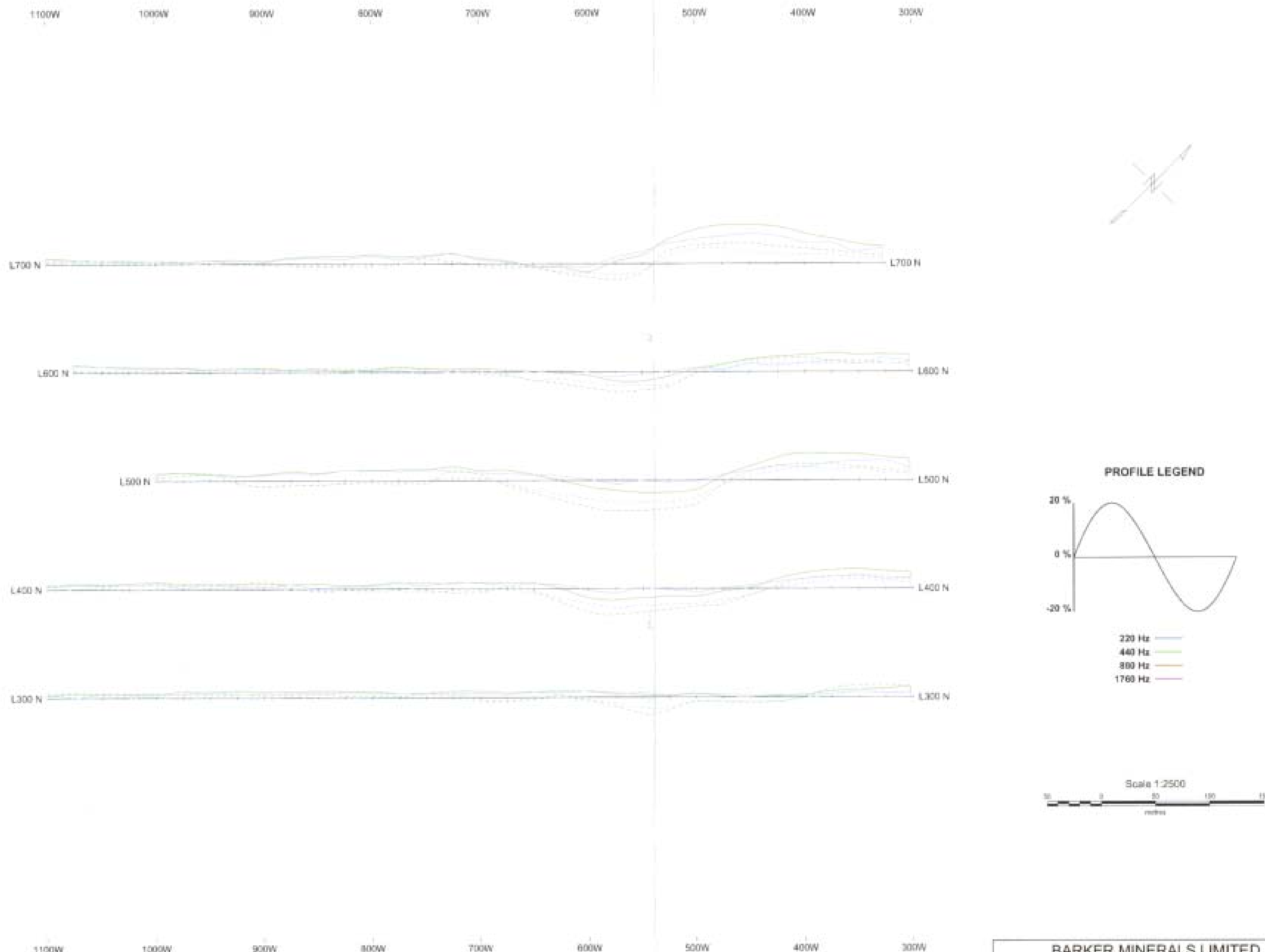


BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

ACE NORTH-N GRID, CARIBOO M.D.
Figure: 3 Date:July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED

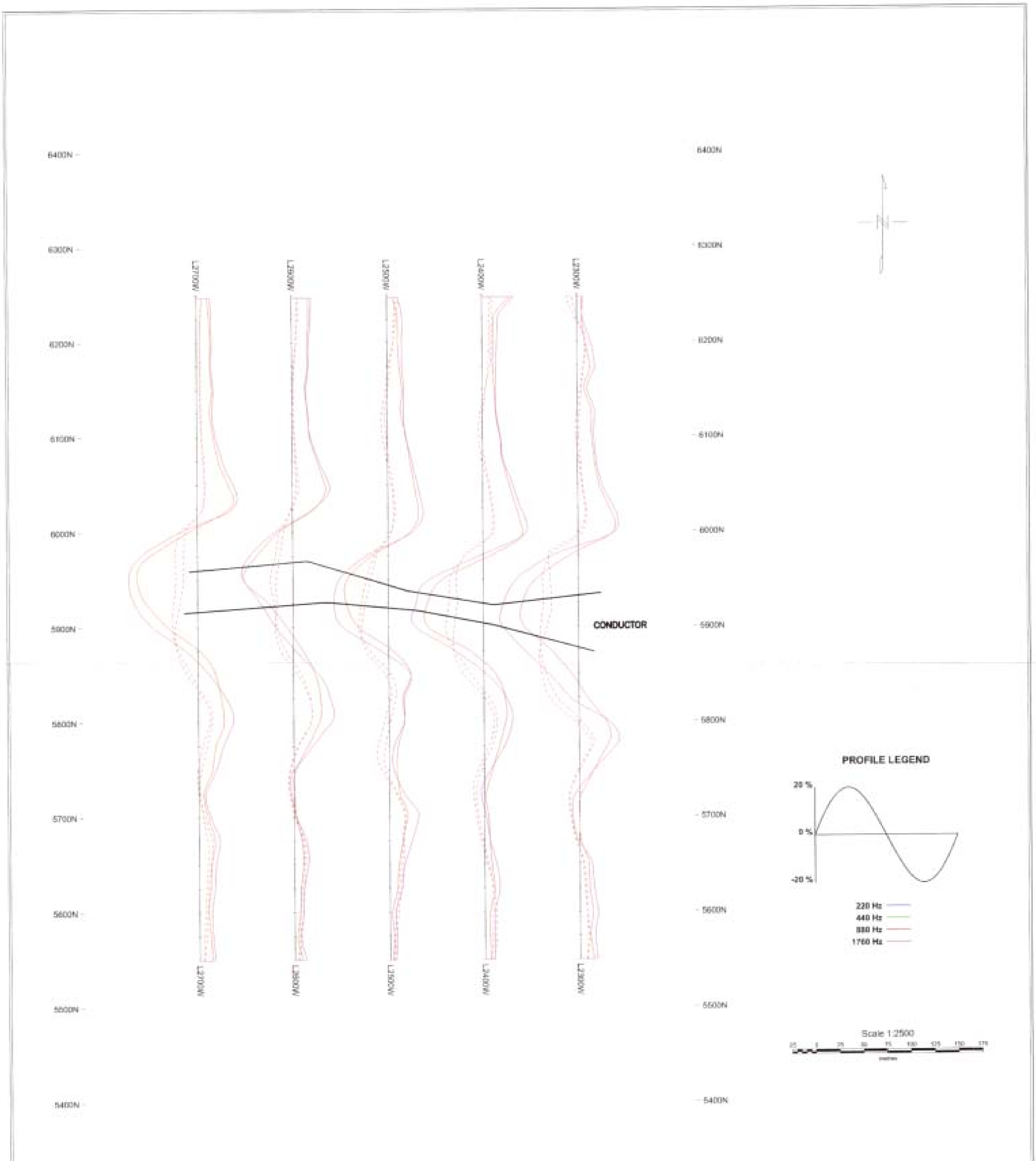


BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

ACE NORTH-N GRID, CARIBOO M.D.
Figure: 4 Date:July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED

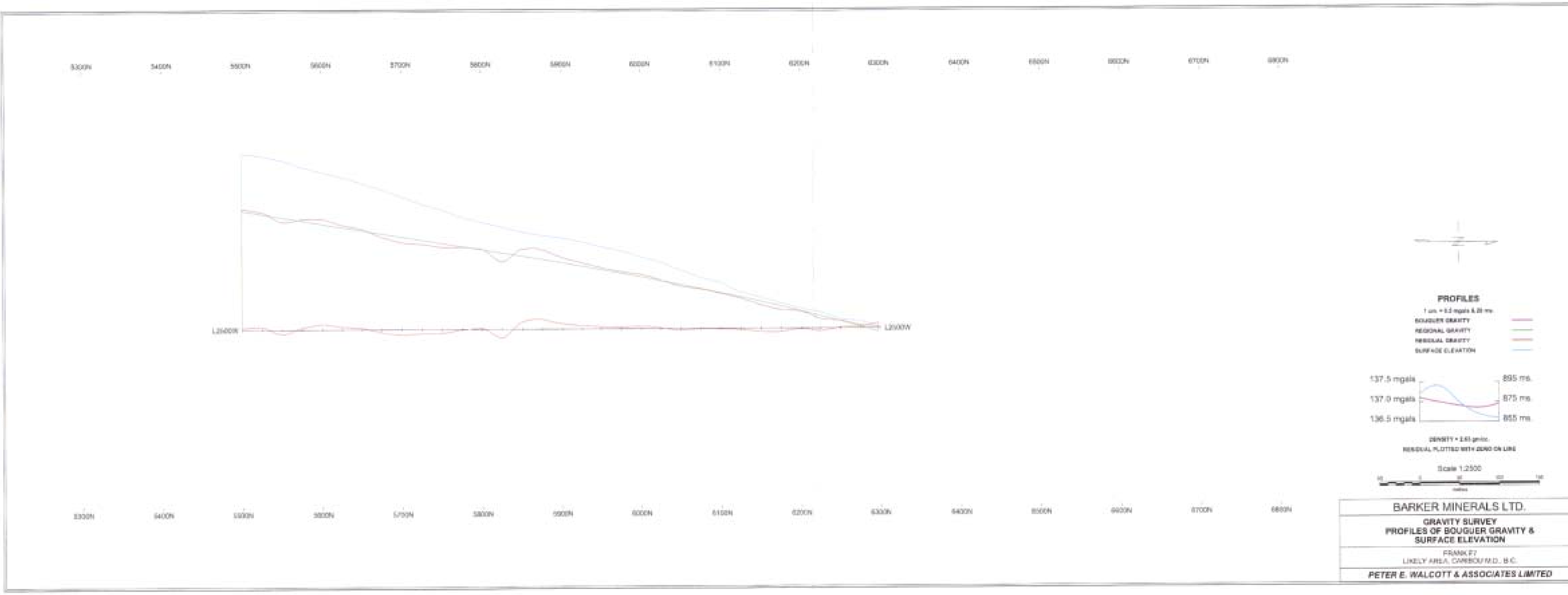


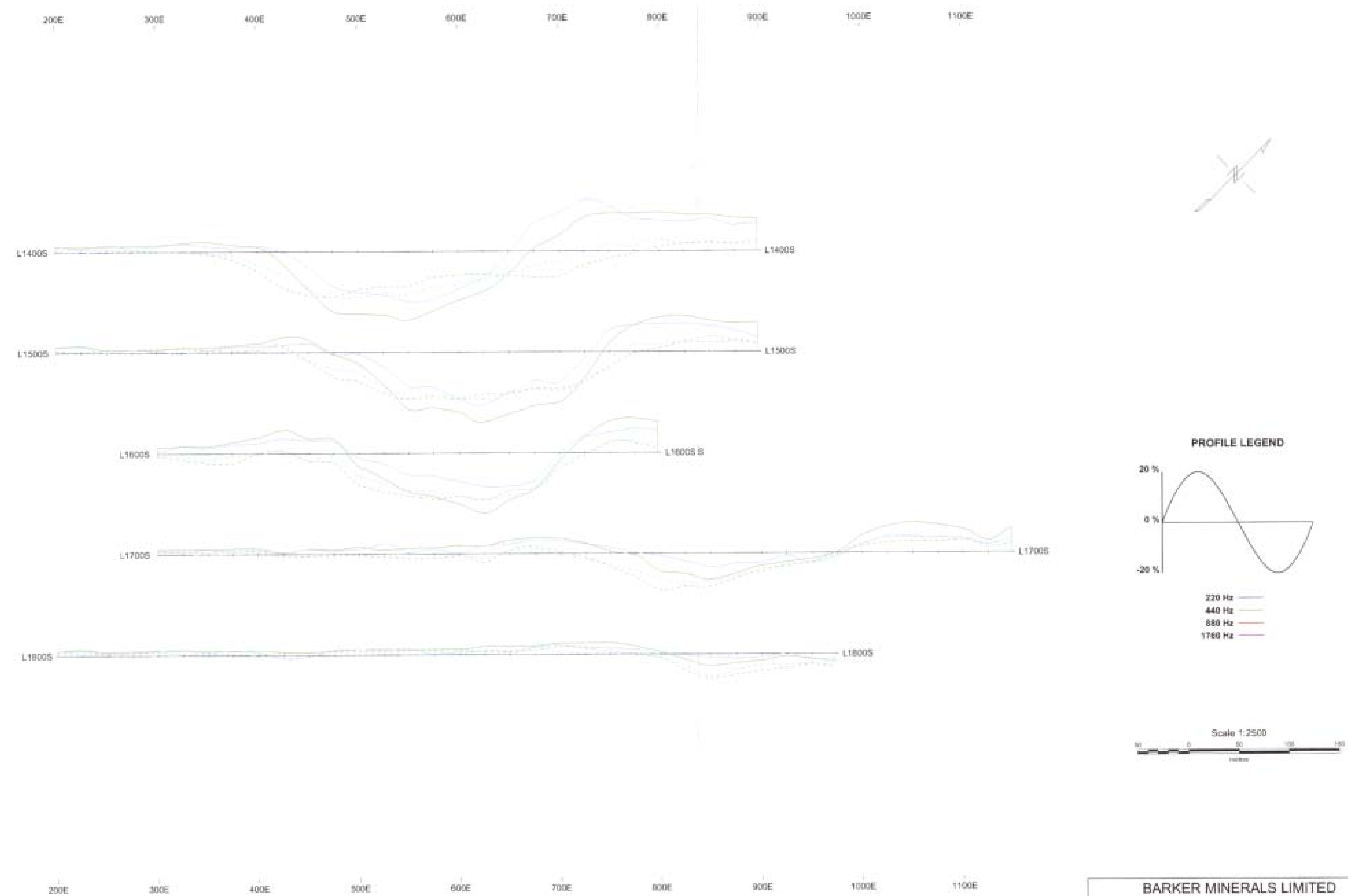
BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

FRANK F7 GRID, CARIBOO M.D.
Figure: Fig 5 Date: July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED


5300N
5400N
5500N
5600N
5700N
5800N
5900N
6000N
6100N
6200N
6300N
6400N
6500N
6600N
6700N
6800N
L2500W
PROFILES
BOUGUER GRAVITY
REGIONAL GRAVITY
RESIDUAL GRAVITY
SURFACE ELEVATION
137.5 mgals
137.0 mgals
136.5 mgals
895 ms
875 ms
855 ms
RESIDUAL PLOTTED WITH ZERO ON LINE
Scale 1:2500
BARKER MINERALS LTD.
GRAVITY SURVEY
PROFILES OF BOUGUER GRAVITY &
SURFACE ELEVATION
FRANK F7
LIKELY AREA, CARIBOU M.D., B.C.
PETER E. WALCOTT & ASSOCIATES LIMITED

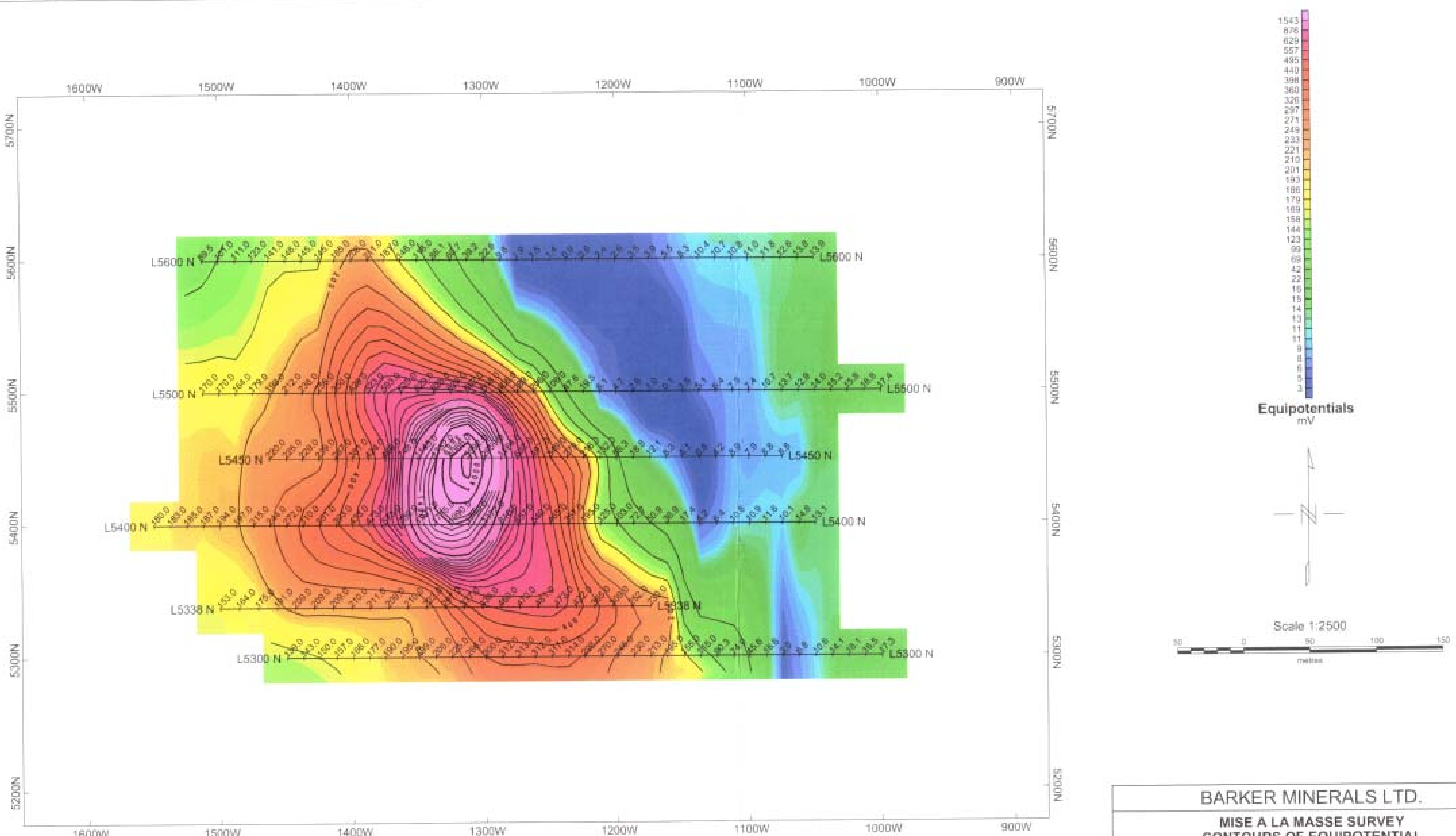


BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

ACE NORTH-S GRID, CARIBOO M.D.
Figure: 2 Date: July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED

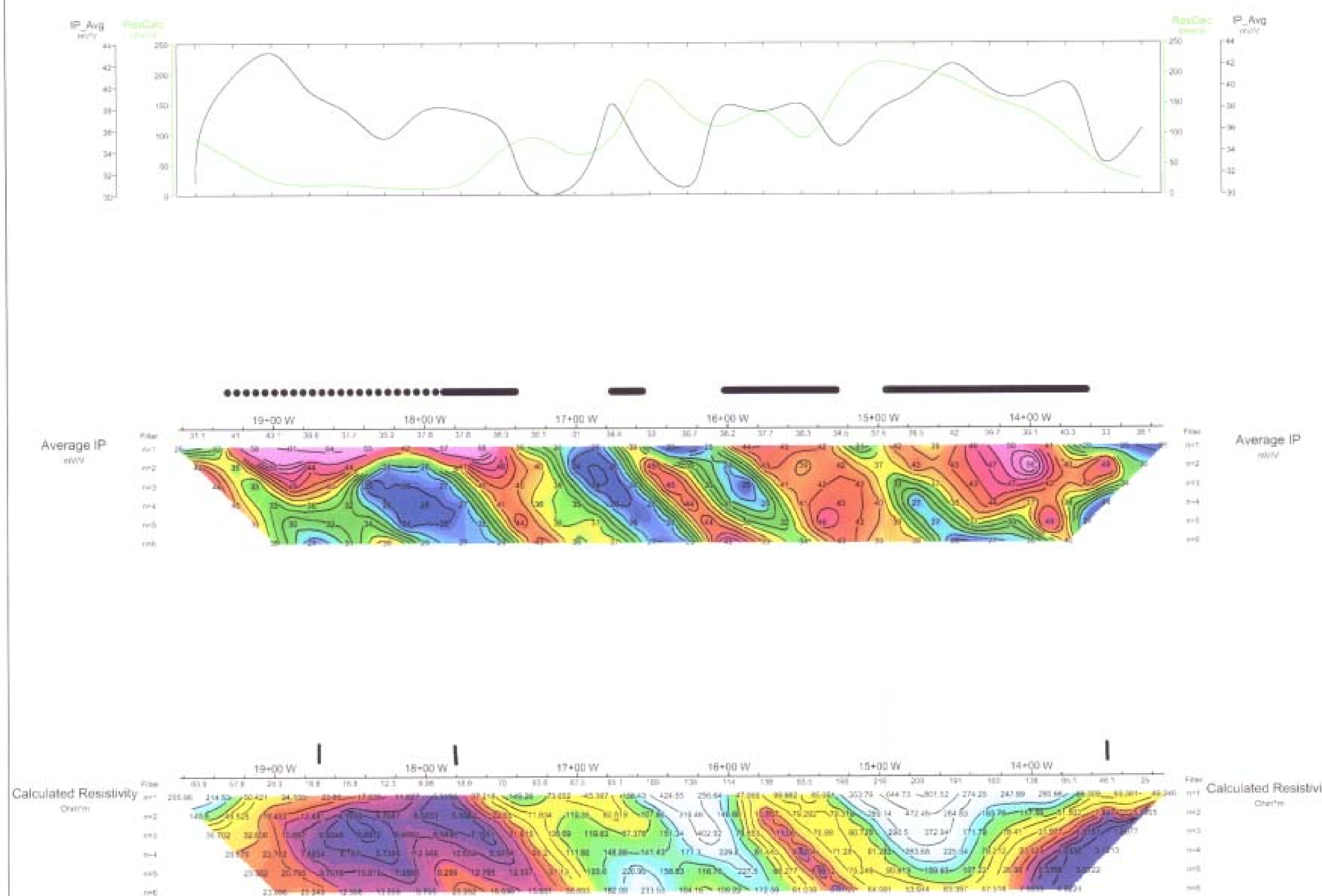


BARKER MINERALS LTD.

MISE A LA MASSE SURVEY
CONTOURS OF EQUIPOTENTIAL
IN MILLIVOLTS

FRANK CREEK SHOWING
CARIBOO M.D., B.C.
FIGURE 15
JUNE 2002

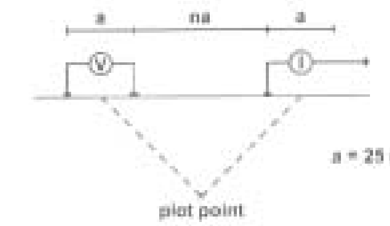
PETER E. WALCOTT & ASSOCIATES LIMITED



58+00 N

Dipole-Pole Array

Filter



Instruments: Iris VIP 4.0 Kw Tx, Iris IP6 Rx.
Frequency: 0.125 Hz.
Operators: M.W., C.B.

Logarithmic Contours 1.5, 2, 3, 5, 7.5, 10,...

INTERPRETATION

Well defined, strong increase in polarization with or without marked decrease in resistivity.

Fairly well defined moderate increase in polarization.

Fairly well defined weak increase in polarization.

Resistivity feature.

Scale 1:2500

25 0 25 50 75 100 125 150

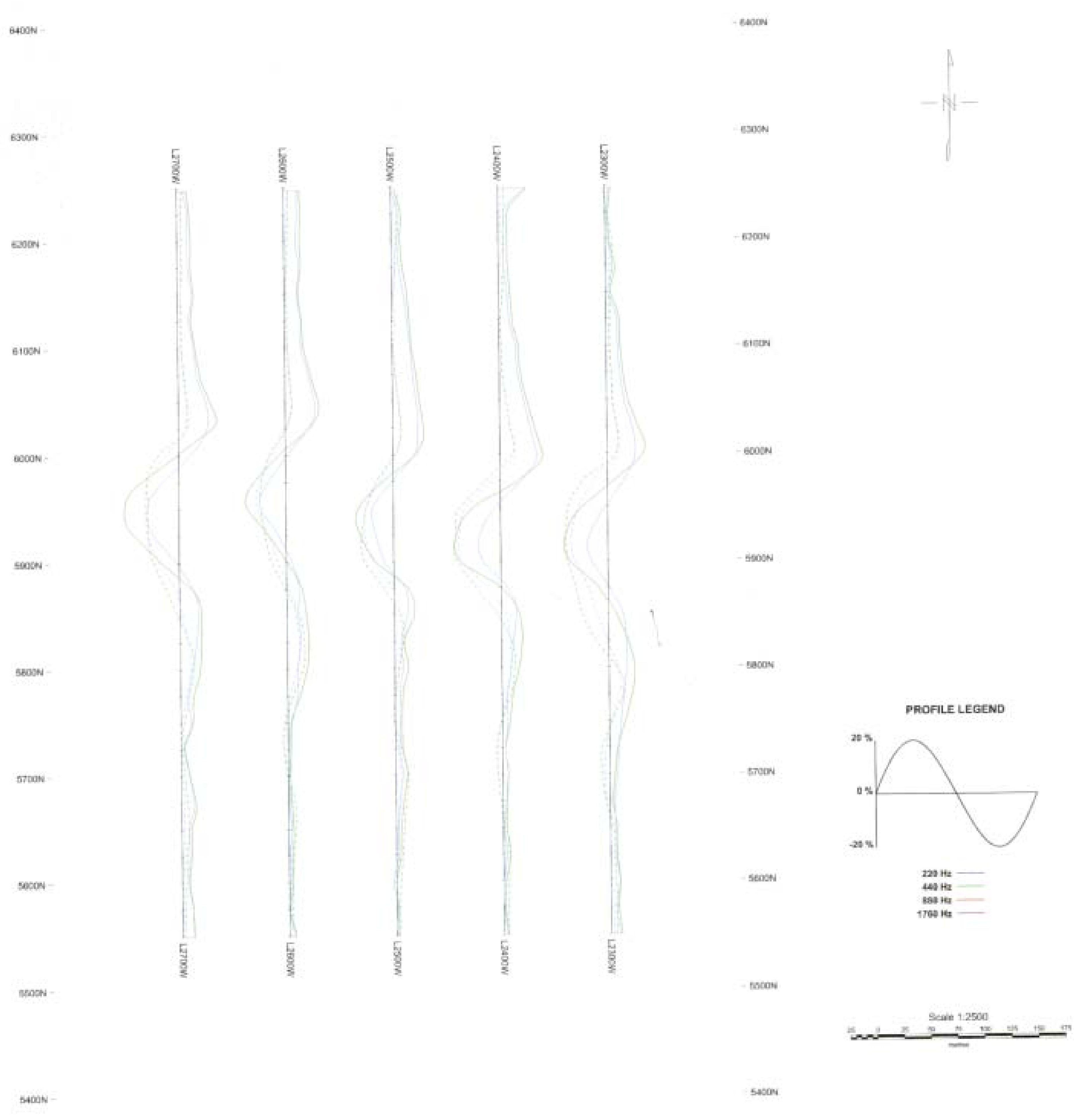
metres

BARKER MINERALS INC.

INDUCED POLARIZATION SURVEY
FRANK GRID, LIKELY AREA
CARIBOO M.D., B.C.

Date: MAY 2002 N.T.S.: 93 A 14
Interpretation: Peter E. Walcott

PETER E. WALCOTT & ASSOC. LTD.

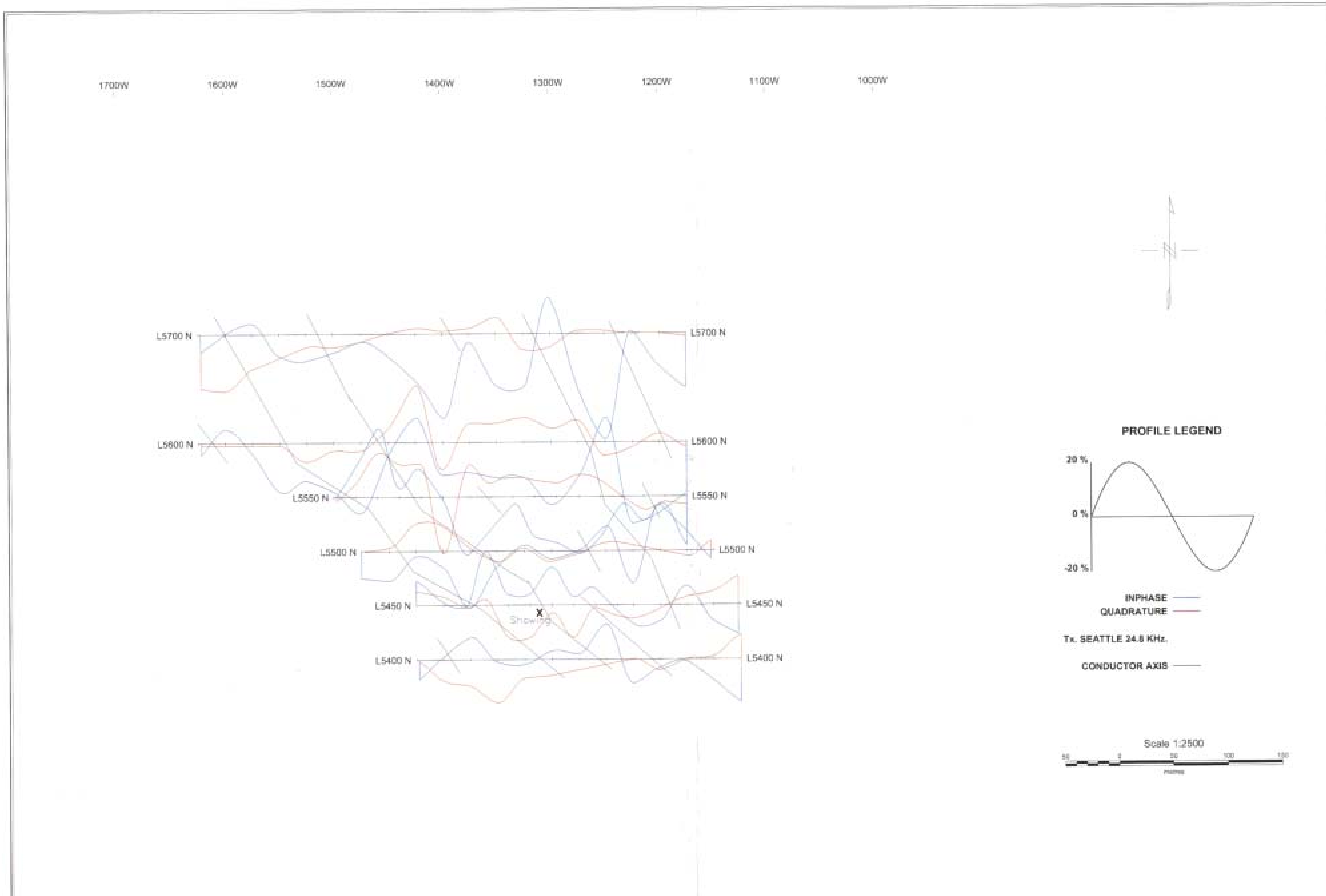


BARKER MINERALS LIMITED

ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & OUTPHASE

FRANK F7 GRID, CARIBOO M.D.
Figure: Fig 6 Date:July 2002

PETER E. WALCOTT & ASSOCIATES LIMITED



BARKER MINERALS LIMITED

VLF ELECTROMAGNETIC SURVEY
PROFILES OF INPHASE & QUADRATURE

FRANK CREEK GRID, CARIBOO M.D.
Figure: 16 Date:JUNE 2002

PETER E. WALCOTT & ASSOCIATES LIMITED

Report on Exploration of the Barker Minerals Ltd. Property, including the Frank Creek and Sellers Creek Road VMS Projects, the Ace VMS and Vein Gold Project and the Quesnel Platinum Project; Cariboo Mining Division, British Columbia, Canada NTS 93 A and 93 B

October 2001 - September 2002

for

Barker Minerals Ltd.,
22117 - 37A Ave.
Langley, B. C. V2Z 1N9

by

Bruce J. Perry, Ph. D., P. Geo., F. G. A. C.
PRO-GEO Exploration and Mining Services Inc.
2301 Skeena Dr.
Kamloops, B. C. V2E 1Y2

Effective Date: October 21, 2002

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Appendices:

Item 3: Summary

The property is located 95 km northeast of Williams Lake in Central British Columbia. This large mineral tenure holding consists of 4,092 mineral claim units (approximately 260,000 acres), of which 291 claim units are owned by Barker Minerals Ltd. and 3,801 claim units are held in trust for Barker Minerals (Figure 1 and Appendix 1). Of these, 453 claim units are subject to a 2% net smelter return royalty in favor of Louis Doyle. The property contains the idle Providence Mine, classified as a 'Past Producer' (BC MinFile 093A 003) of silver, lead, zinc and gold. The property also contains the Cariboo (a.k.a. Maybe) Prospect, which is classified as a 'Developed Prospect' (BC MinFile 093A 110). This is reported to be a lead-zinc (Zn-Pb) replacement-style deposit estimated to contain approximately 400,000 tonnes at an estimated grade of 4% Zn+Pb, using a 1% Zn+Pb cutoff. The property contains the Ace VMS Prospect, which was discovered during 1993 by Louis Doyle, President and CEO of Barker Minerals and is host to what has been described by BC Geological Survey geologists as Besshi-type volcanogenic massive sulphide (VMS) mineralization and auriferous (gold-bearing) quartz veins (BC MinFile 093A 142). The property contains several known bedrock mineral occurrences, classified as 'showings' by the BC Geological Survey, including the Frank Creek VMS showing (BC MinFile 093A 152), the Sellers VMS showing (BC MinFile 093A 131), the Big silver-lead-gold showing (BC MinFile 093A 151), the Comin Throu Bear lead-zinc-silver showing (BC MinFile 093A 148), the Peacock gold-bearing quartz vein showing (BC MinFile 093A 133), the Maud alkalic porphyry copper-gold showing (BC MinFile 093A 119) and the Trump silver-lead showing (BC MinFile 093A 154). Both the Frank Creek VMS showing and Sellers Creek Road (SCR) VMS showing were discovered by Louis Doyle, President of Barker Minerals Ltd.

The eastern half of the property contains three VMS exploration project areas, the Ace, Frank Creek and SCR areas, each of which contain multiple exploration targets as indicated by geochemical, geophysical and geological data and which have been the focuses of Barker Minerals' exploration programs during the period 1993 to the present. The western half of the property contains the mineral claims hosting Barker Minerals' Quesnel Platinum Project (Figure 2).

Within the Ace project area (Figures 1, 4, 4a, 4a-1, 4a-2, 4b, 4c, 4d, 4e and 6b), surface geological, geochemical and geophysical surveys and two episodes of drilling [7 DDH (1260 m) in1998; 5 DDH (645 m) in 2002] have defined a belt of metamorphosed and deformed, felsic volcanic rocks containing massive and stringer sulphide mineralization, within which are anomalous concentrations of gold, silver, copper, lead and zinc. The belt is open along strike in both directions. Geophysical surveys have defined another major target located to the southeast of this belt in an apparently outcropless area containing encouraging soil geochemistry. Further exploratory trenching and drilling are recommended on these targets.

Within the Frank Creek project area (Figures 1, 5a, 5b and 5c), a massive sulphide layer 1.2 m thick outcrops in a zone of overturned volcanic and sedimentary rocks. During 2002 a series of trenches were excavated in order to test several targets in and adjacent to the F-1 Target Area at the Frank Creek VMS project. Five exploratory trenches were excavated over a total distance of 289 metres up to 70 metres southeast of the discovery outcrop. Trench TR-BW-05 exposed the dark quartz eye phyllite host unit within which were found several small massive sulphide lenses within the projected strike extension of the mineralized zone exposed at the discovery outcrop. TR-BW-04 exposed copper stringer mineralization 60 metres southwest of the F-1 occurrence. Along with the mineralized exposure in TR-BW-10 excavated some 375 metres to the NW of the discovery outcrop, this exposure extends the known strike length of the mineralized zone to approximately 425 metres. The zone is open to potential extensions in both directions to the north and south and at depth. According to Wild (2002), former Chief Geologist of Goldstream Mine, near Revelstoke, B. C., the geological setting, mineralization and host rocks are all remarkably similar to the Goldstream Mine mineral deposit, which produced more than 2 million tonnes of ore at a grade in excess of 4.0% copper and 2.2% zinc.

Drill core from the initial exploratory drilling program [6 ddh (813 m)] at the Frank Creek project area contains intervals of Cu-Zn-Pb (+/- Au, Ag) massive sulphide mineralization that are significant examples of ore formation processes having occurred on the property. The mineralizations encountered in the drill core are similar to that exposed at the discovery outcrop where the discovery outcrop massive sulphide layer has been further exposed (3.5m in length and 1.5 metre wide) by trenching, and the local area mapped in detail (Wild, 2002). The concentrations of metals from grab samples of the outcrop massive sulphides ranged up to 4.4% Cu, 8.2% Zn, 1.1% Pb, 14.8oz/t Ag and 854 ppb Au. A chip sample across 5.77 metres exposed width assayed 0.82% Cu, 0.25 % Zn, 0.21% Pb and 44.3 g/t Ag including a .77 metre exposed width which assayed 2.1% Cu, .34% Zn, .11% Pb and 69 ppm Ag.

This Besshi-type VMS polymetallic mineralization also occurs in drill core in significant intervals (up to 0.4 metres) and contains significant concentrations of zinc, copper, silver, lead and gold (up to 3.4% Zn, 2.1% Cu, 2.8 oz/t Ag, 0.53% Pb and 746 ppb Au) within larger weakly mineralized units up to 52 metres wide in drill core intercepts that contain widespread disseminations of these metals. Several significant, combined geophysical and geochemical anomalies are present, one of which is situated within the area in which the massive sulphide outcrop occurs. Further exploratory trenching and drilling are recommended in order to test these specific targets for economic mineralization and in order to further define the extent of the mineralized zone.

The SCR project (Figures 1, 11a and11b) area contains semi-massive sulphide mineralization in altered volcanic rocks. This project area also contains coincident base-metal soil anomalies and HLEM geophysical anomalies in an area of sparse outcrop. A Maxmin geophysical survey comprised of 4.2 line kilometers was completed during the 2002 field season. In areas of geophysical and geochemical anomalies, prospecting was successful in discovering float boulders which assayed as high as 17.3% Zn and 6.4% Pb (Table 4) Further surface exploration including trenching and bedrock sampling in this area is recommended, to be followed by initial exploratory drilling, if warranted.

The Cariboo Zn-Pb deposit (Figures 8b) is reported to be comprised of replacement style Zn-Pb mineralization hosted in carbonate rocks of the lower strata of the Cariboo terrane. Diamond drilling conducted during the1980's outlined a 400,000 tonne deposit grading 4.0% Zn+Pb (BC MinFile 093A 110). Further surface mapping should be conducted in this area in order to help gain an understanding of the deformation history of the deposit and the potential for extensions of the known zone. Compilation of all relevant data and limited diamond drilling is recommended in order to confirm the previous operator's drilling and in order to further define and investigate the size and economic potential of this deposit, which is open in both directions along strike and at depth.

The western part of the property (Quesnel Platinum Project area) was staked for its platinum group element (PGE) potential (Figures 1 and 2). It contains zones of anomalous and intense copper concentrations in mafic volcanic rocks and may contain some of the mafic to ultramafic source rocks for some of the platinum group minerals (PGMs) recovered from the predominantly gold-bearing placers associated with the Quesnel River and its tributaries.

Further geochemical and geological surveys are recommended in order to explore for bedrock PGE mineralization and in order to explore the zone of mafic volcanic rocks containing anomalous concentrations of copper.

Exploration work conducted by Barker Minerals Ltd. to the present date has been conducted under the authority of Mineral Exploration and Reclamation Permit MX-10-155, which may be modified by amendments in order to facilitate future work such as that recommended within this report.

The estimated costs of the Phase 1 and Phase 2 programs are $1,780,000 and $4,000,000 respectively.

Item 4: Introduction and Terms of Reference

This report was prepared for Barker Minerals Ltd., as an annual professional geo-scientist's report regarding exploration work conducted on its mineral exploration property in the Cariboo Mining Division of the Province of British Columbia during the period September 2001 through September 2002. Dr. Perry, P. Geo., F. G. A. C., the author of this update report, has been involved with the project intermittently since August, 2000, conducting geological reconnaissance, rock sampling, property inspections, report writing and acting as the Company's independent, qualified person per regulatory requirements, as required.

Work on the property to date includes geological mapping, stream-sediment and soil surveys, ground magnetometer and EM studies (VLF-EM, HLEM), airborne magnetometer and EM studies, prospecting, and, in some areas of potential economic interest, trenching and diamond drilling.

The Ace Prospect mineralization and Frank Creek and SCR mineral showings are of the VMS type. The Cariboo Prospect hosts a replacement style Zn-Pb deposit in carbonate rocks of the Cariboo terrane. Scattered stream-sediment anomalies situated within the claims have not been followed up in detail. The Quesnel Platinum Project areas may contain some of the source mafic and ultramafic rocks for PGE minerals occurring in numerous placer platinum occurrences associated with the Quesnel River and its tributaries. These claims also host mafic volcanic rocks that in one area contain anomalous concentrations of copper.

Item 5: Disclaimer

The author Perry has relied on the reports of J. Payne, Ph. D., C. Wild, P. Geo. and P. Walcott, P. Eng., which describe geological and geophysical exploration, respectively, conducted on the Company's property during the subject period. These reports are those listed within the References section of this report.

Item 6: Property Description and Location

The mineral exploration property consists of 4,092 mineral claim units, being approximately 260,000 acres or 105,222 hectares (Figure 1). The mineral claims comprising the property are owned by and registered in the name of Barker Minerals Ltd. or Louis Doyle in trust for Barker Minerals Ltd. (Appendix 1). Of these, 453 claim units are subject to a 2% net smelter royalty in favor of Louis Doyle. The property contains 16 mineral exploration project areas, some of which are currently under active exploration, including the Ace, Frank Creek, Sellers Creek Road (SCR) and Quesnel Platinum project areas. The mineral claim tenure numbers and expiry dates of the mineral claims comprising Barker Minerals' exploration property are listed in Appendix 1.

The centre of the property is situated 95 km northeast of the city of Williams Lake, B.C., the nearest supply center, and 34 km northeast of Likely, B. C., the nearest settlement (Figure 1).

Exploration work conducted by Barker Minerals Ltd. to the present date has been conducted under the authority of Mineral Exploration and Reclamation Permit MX-10-155, which may be modified by amendments in order to facilitate future work such as that recommended within this report. The author has reviewed recent correspondence from the local BC Inspector of Mines indicating to Barker his satisfaction with Barker's progressive reclamation of trench and drill sites, and indicating also his satisfaction with the amount of the current reclamation bond Barker has supplied, without need for any further increase in the bond in respect to additional work planned for the remainder of 2002 (pers. comm., K. McDonald, Inspector of Mines, September 3, 2002).

Item 7: Accessibility, Climate, Local Resources, Infrastructure and Physiography

Williams Lake is an intermediate-sized city, which is served by Highway 97, the B.C. Railway, a major hydroelectric power grid and a modern airport. By road, Likely is 65 km northeast of 150 Mile House on Highway 97. Access to the Ace, Frank Creek and SCR exploration areas is *via* gravel logging roads bearing northeast from Likely. The distances from Likely to the main showings are as follows: Ace, 45 km; Frank Creek, 25 km; SCR, 22 km. Driving time to the Ace prospect from Likely is forty-five minutes. Access to the Quesnel Platinum project is mainly *via* gravel logging roads southeast from Quesnel, a distance of 25 km. In Likely, Barker Minerals maintains a property that includes a house, a bunkhouse, a workshop and a few tents. The house serves as a field office.

The property is situated in the central part of the Quesnel Highland between the eastern edge of the Interior Plateau and the western foothills of the Columbia Mountains. This area contains rounded mountains that are transitional between the rolling plateaus to the west and the rugged Cariboo Mountains to the east. Pleistocene and Recent ice sheets flowed away from the high mountains to the east over these plateaus and down to the southwest (Cariboo River), west (Little River) and northeast (Quesnel Lake), carving U-shaped valleys. The elevation ranges from 700-1650 m.

Precipitation in the region is heavy, as rain in the summer and snow in the winter. Drainage is to the west *via* the Cariboo, Little and Quesnel Rivers to the Fraser River. Quesnel Lake, the main scenic and topographic feature in the region, is a deep, long, forked, glacier-carved lake with an outlet at 725 m elevation. Vegetation is old-growth spruce, fir, pine, hemlock and cedar forest in all but the alpine regions of the higher mountains (mainly above 1400 m elevation). Weldwood has been actively logging fir, spruce and pine in the area (e.g., logging clear-cuts shown in aerial photo Figure 5c), principally during winters, and has provided outlines of existing and planned roads and cut-blocks in and near the project areas.

Item 8: History

Gold was discovered in the Barkerville-Wells area in 1858. Historical production totaled 3.7 million troy ounces, as 1.9 million ounces from placers and 1.8 million ounces from 2.7 million short tons of underground ore. The historic Bullion Pit near Likely produced 175,700 ounces of gold from 200 million tons of gravel and about 1/100th as much platinum. Gold and platinum were reported in similar proportions in placers in the Frank Creek area.

Much of the lode and placer gold production from the Wells and Barkerville areas occurred in the eastern part of the Barkerville terrane, where most of the important present day lode gold prospects also occur. At the historic mines, the strata trend 315 , dip 45 NE and are cut by north to northeast-trending normal faults dipping 60 E. The zones of economically important quartz veins are contained in graphite-bearing layers near a contact with carbonate-bearing layers. Two sets of quartz veins contain pyrite and gold: transverse veins striking 030 and diagonal veins striking 070 ; these contained 15 -25% pyrite and 0.3-0.5 oz/st Au. Two other sets of quartz veins do not contain pyrite or gold. Pyrite in altered wall rock contained proportionally less gold. Replacement bodies of auriferous pyrite in carbonate rocks (mainly in the Island Mountain Mine north of Jack of Clubs Lake) accounted for nearly 1/3 of the lode gold mined. Other common sulphides in the mines are arsenopyrite, galena, sphalerite, cosalite (PbBiS), scheelite and bismuthinite (BiS_2). Free gold commonly occurs with nests of cosalite. Gangue minerals are quartz and ankerite/siderite. Wallrock alteration minerals are ankerite and sericite.

Frank Creek area

Frank Creek has hosted sporadic placer mining since the turn of the century, the latest periods being 1984 to 1986 and 2000 to 2001. During early placer mining, several massive sulphide boulders containing up to 9.3% Zn+Pb (Formosa 1989) were found in a thin, clay-rich layer situated near the base of the alluvial material where Frank Creek canyon opens onto a delta.

During 1988, Formosa Resources Corp. and Rio Algom Mines conducted grid-controlled soil sampling, VLF-EM geophysical surveys and limited geological mapping. As a result of this work, they identified several coincident geophysical and soil geochemical anomalies. A trench excavated on a coincident Cu-Zn-Ag soil anomaly encountered rusty soil and lenses of ferricrete containing pyrite-rich fragments that contain also minor chalcopyrite, galena and sphalerite. On the slope above the above-mentioned alluvial massive sulphide boulders, 22.4 km of grid lines were cut and soil sampling was conducted at 25-m intervals, yielding some anomalous Cu, Pb, Zn and Ag concentrations but no outcropping bedrock mineralization.

During 1991, helicopter-facilitated magnetic, electromagnetic (EM) and radiometric surveys by Rio Algom identified seven areas containing anomalous EM signatures in the Frank Creek area, F-1 to F-7 target areas (Figures 5a and 5b).

During 1996, two, vertical, percussion holes were drilled along the D-Road switchback near the center of the F-1 target area by R. Yorston, a previous operator. They yielded poor recovery, mainly of very fine particles and minor amounts of rock chips of black and green phyllite. Both holes returned anomalous Cu, Pb, Zn, Ag and Au concentrations in the upper 18.3 m, with the best interval being 3.04 m averaging 1770 ppm Cu, 750 ppm Pb, 2970 ppm Zn, 4.8 ppm Ag and 790 ppb Au.

During the period covered by this report, the Company conducted geological and geophysical exploration work, including trenching, rock sampling, localized detailed mapping at the Frank Creek VMS discovery outcrop, diamond drilling and assaying (items 11, 12, 13 this report).

Cariboo Zn-Pb Deposit

Barker Minerals' Cariboo Prospect, also known as the Maybe Prospect, was explored during 1987 by Gibraltar Mines Ltd. The prospect contains three main stratiform lenses of ankerite, quartz, sphalerite, galena and minor pyrite enclosed in limestone-rich strata of probable Middle Devonian age. The large Zn/Pb ratio, moderate silver content and low gold content are similar to those of many carbonate-hosted, replacement Zn-Pb deposits, including those in the Early Cambrian platform carbonates of the Kootenay arc. Limited bulldozer trenching and 21 diamond drill holes by Gibraltar Mines resulted in an in-house mineral inventory estimate of 400,000 tonnes at a grade of 4% Zn+Pb. This is not an appropriate category and does not comply with the requirements of current securities legislation. The author has recommended that limited confirmatory fieldwork and limited confirmatory drilling be done in order to define and categorize the known mineralization to current day industry and regulatory standards.

Quesnel Platinum Project areas

Some placer deposits associated with the Quesnel River and some of its tributaries emanating from the north and east contain potentially significant concentrations of platinum group metals (Rublee, 1986 and Figure 2). The highest concentration was obtained from a pan concentrate sample collected from Twenty-Mile Creek that assayed 2195 g/t Pt, 2210 g/t Pd and 1440 g/t Os. In this concentrate, the platinum group minerals (PGMs) were found as minute metallic grains within larger grains of magnetite and chromite.

A regional stream survey conducted by the BC Geological Survey during 1980 (GSC Open File 776), with results re-analyzed during 1997 by neutron activation (BC Geological Survey; BC RGS 50), showed anomalous concentrations of Cu, Ni, Cr, Co and Au in and around areas of ultramafic rocks that are situated partly along the Eureka fault and partly within the Quesnel terrane. An airborne magnetometer survey indicated several strong anomalies associated with ultramafic rocks located along the Eureka fault and at known locations of magnetite-bearing mafic and ultramafic rocks in the Quesnel terrane (Dome Exploration, 1981). During 1984, helicopter-facilitated multi-frequency EM, VLF-EM and magnetometer surveys were conducted in the Maud Lake, Victoria Creek and Trumph areas and many EM and magnetic anomalies were detected and recommended for further geological and geochemical investigation (BC Assessment Report #12780). During the late 1980s, QPX Minerals drilled two holes into ultramafic bodies at its Maude Lake property. Both holes, a few hundred metres apart, contained geochemically anomalous concentrations of Pt up to 35 ppb (1988, BC Assessment Report #17598). At the bottom of one of the drill holes was a 10-cm intersection of chalcopyrite-pyrite massive sulphide for which no assays are available. Mandella Resources conducted ground geophysical and geochemical soil surveys west of this area in the mid-1980's and identified vlf-EM geophysical anomalies and Pb, Zn, Cu and Au soil geochemical anomalies, which in some instances were coincident (BC Assessment Report #14816). Some of the PGM contents of the above mentioned Quesnel River placer deposits are probably derived from ultramafic rocks situated along the Eureka fault between the Quesnel terrane and the Barkerville terrane and from mafic and ultramafic rocks in the Quesnel terrane. Barker Minerals' claims in this area were staked because of potentially favorable geology, the recent dramatic increase in the price of PGMs and the apparent lack of previous PGE exploration in this potentially favorable area.

Reconnaissance geological, geochemical and airborne geophysical surveys were conducted over portions of the area by several companies in the 1980s, mainly during exploration for porphyry copper deposits. Some of these studies identified Cu-Pb-Zn-Ag soil anomalies that are coincident with HLEM-indicated conductors which have not yet been tested by trenching or drilling.

Item 9: Geological Setting

The regional geology was described by L.C. Struik (1988) and is shown updated in Figure 3 (after Ferri, 2001). The Barkerville terrane is considered to be the northwest extension of the Kootenay terrane, which to the southeast overlies the Monashee metamorphic core complex, a large uplifted mass of high-grade paragneiss, quartzite and marble. The properties are on the flank of the northern, unexposed portion of this core complex. Northwest from the North Arm of Quesnel Lake the characteristic metamorphic minerals change from sillimanite through staurolite-kyanite, almandine garnet and biotite to chlorite northwest of the Ace claims. The garnet isograd runs northerly across the east-central part of the Ace group, while that of biotite is 30 km further northwest. Historic mines near Wells and Barkerville are in rocks of the greenschist facies. The age of both deformation and metamorphism is regarded as Mid-Jurassic, which is interpreted as the time of collision of the North American plate to the east with a group of island arcs to the west. In the Little River area, four geological terranes are represented, most of which are dominated by marine sedimentary or metasedimentary rocks (Figure 3).

Barkerville Terrane

Most of the property area is underlain by marine strata of the Barkerville terrane, whose age is classified broadly as Late Proterozoic to Mid-Paleozoic. It is categorized by the Geological Survey of Canada as a subdivision of the Kootenay terrane. The region was deformed by intense, complex, in part isoclinal folding and overturning that produced an intimate interlensing of impure quartzite, siltstone, ankeritic dolomite, pelite and amphibolite. These rocks are cut by dikes and sills of metamorphosed diorite. Locally, stronger shear deformation produced mylonitic textures.

The northeastern third of this terrane is the main zone of economic interest in the Cariboo district. Struik described it as "gold-enriched", because it contains the historic Wells and Barkerville mines and the Cariboo Hudson deposit, 39 km and 18 km northwest of the Ace project area, respectively. This zone contains olive and grey micaceous quartzite and phyllite, amphibolite, marble, meta-tuff and meta-diorite sheets or sills. These descriptions are compatible with the rock types on the Ace project area, although the latter contains more metamorphosed felsic/intermediate volcanic rocks. Stratigraphic tops are unknown. The Barkerville terrane is cut by the Mid-Devonian Quesnel Lake gneiss (350 Ma), a coarse grained, leucocratic, biotite granitic gneiss with megacrysts of potassium feldspar. The main body of gneiss is 30 km long by 3 km wide and is elongated parallel to the eastern border of the Intermontane belt. Its contacts are in part concordant with, and in part perpendicular to, metamorphic layering. The Barkerville terrane hosts folded, sill-like masses up to 300 m thick of gneissic meta-diorite (400 Ma) and contains post-metamorphic anatectic pegmatite (86 Ma), particularly in a high-grade metamorphic aureole northwest of the North Arm of Quesnel Lake.

Cariboo Terrane

The northeastern part of the Little River area is underlain by marine peri-cratonic sedimentary strata of the Cariboo terrane. The Cariboo terrane consists mainly of limestone and dolomite with lesser siliceous, clastic, sedimentary rocks and argillite. Some geologists believe that the Cariboo terrane is a shallow, near-shore facies and the Barkerville is a deeper, offshore facies of the same erosion-deposition system. No rifting is suspected between the Cariboo terrane and the North American continent, in contrast to that between the Barkerville terrane and the North American continent.

The Cariboo and Barkerville terranes are separated by the regional Pleasant Valley thrust fault, which dips northeast moderately to steeply. It is reported by Struik (1988) to have moved the Cariboo block from the east over the Barkerville block along a strike length of over 100 km. In the map area, the fault cannot be found, suggesting that much of the movement attributed to it may have occurred by shearing in a broad zone along the "contact" between the two terranes.

Some of the carbonate layers in the lowest part of the Cariboo terrane (or upper part of the Barkerville terrane) are enriched in zinc and lead. Since the 1970's, preliminary exploration on stratiform Zn-Pb targets has been conducted in this area over a strike length of 23 km from the vicinity of the head of the North Arm, *via* Maeford Lake to the Cariboo (Maybe) prospect.

The Cariboo terrane was cut by the Jurassic-Cretaceous Little River stock, a medium-grained granodiorite grading to quartz monzonite. A normal fault along its southwest side (Little River fault) dips east and extends southeasterly to Limestone Point, on the western side of the North Arm of Quesnel Lake. It intersects, and in some literature has been confused with, the Pleasant Valley thrust. It moved chlorite-biotite metamorphic grade strata of the Cariboo terrane eastward to rest against staurolite-kyanite metamorphic grade strata of the Barkerville terrane.

Quesnel Terrane

A small southwestern portion of the Little River area is underlain by the Late Triassic to Early Jurassic, allochthonous Quesnel terrane. It was accreted to the North American continent, in part by subduction and in part by obduction. The Eureka thrust fault marks the boundary between the Quesnel and Barkerville terranes as well as that between the Intermontane and Omineca physiographic belts. The terrane is partly submarine and partly subaerial, consisting of volcanic and volcaniclastic rocks and co-magmatic intrusions, with minor carbonate lenses and related sedimentary rocks. Regionally, it hosts many important mineral deposits, mainly of Cu and Cu-Au, such as Highland Valley, Craigmont, Copper Mountain, QR and Mt. Polley. The Bullion Pit, from which 175,700 oz. of placer gold were produced, is near Likely just on the west side of the boundary between the Barkerville and Quesnel terranes.

Slide Mountain Terrane

Rocks of the allochthonous, Devonian to Late Triassic, Slide Mountain terrane underlie a very small part of the Little River area. Portions of these rocks were obducted, while others were subducted during collision of an oceanic plate with the continent. It is exposed east of Wells and Barkerville as the upper plate overlying the generally low-angle Pundata thrust fault. This fault it is nearly vertical where it crosses the southwestern part of the Little River area. Small slices of mainly mafic volcanic rocks and alpine-type ultramafic rocks of the Slide Mountain terrane occur in and parallel to the Eureka thrust. Minor lithologies include chert, meta-siltstone and argillite.

Ace project area

The Ace project area and vicinity are underlain by the Barkerville terrane that, except for its easternmost part, is believed to be Late Proterozoic. In spite of the amount of work done prior to 1998, the Ace project area had not been mapped in detail geologically, partly because of scarcity of outcrop. Most known outcrops are situated in creeks, in road cuts or in several of the trenches excavated by Barker Minerals during 1996 and 1997.

During the fall of 1998, low water levels in the major rivers allowed mapping of extensive bedrock exposures in their floors and walls. Most of the outcrop in the Little River is exposed southeast of the bridge on Colleen Road. West of there, the river valley is mainly covered by alluvial deposits with scattered small outcrops, mainly at major bends. The northern slope of Barker Mountain is virtually devoid of outcrop from just south of the main trench area to the zone of abundant outcrops near the crest of the mountain. The diamond-drilling program in 1998 gave the first detailed stratigraphic sections across the main zone of felsic volcanic rocks. Many geological questions remain unanswered, including the existence and location of the Pleasant Valley thrust fault and two northeast-trending, cross-faults (GSC-1 and GSC-2) that were reported to offset the Pleasant Valley thrust fault (Struik, 1988).

The geology of the Ace project area is shown in Figure 4a. A fairly continuous stratabound interval of plagioclase-rich rocks, interpreted as metamorphosed and altered felsic volcanic rocks, extends in an east-west direction for 7 km across the centre of the Ace project area from near the "J" Road at the west to Little River at the east. Because of very sparse outcrop west of the trenches, the thickness of felsic volcanic rocks in this area is unknown. In drill cores, the true thickness of the felsic unit is up to 80 metres. To the east, where the main band crosses Little River, the felsic volcanic unit is several metres thick. In the hanging wall (to the northeast) is a zone of quartz-muscovite-chlorite-leucoxene meta-sedimentary rocks containing intervals of bedded quartzite and quartz-rich schist. In the footwall (exposed mainly in drill core) are similar sedimentary rocks, but mainly lacking leucoxene.

North of Little River along a logging spur that extends east from the J road is a band up to 50 m thick of similar, altered felsic volcanic rocks containing disseminated lenses of pyrite or pyrrhotite. These grade upwards into quartz-muscovite-(chlorite) schist with local intervals of quartzite. Further up section to the northeast (probably in the Cariboo terrane) is a zone containing black, siliceous argillite to quartzite and interlayers of banded marble.

The felsic volcanic unit is dominated by plagioclase with much less abundant amounts of muscovite, biotite and quartz. It commonly contains 0.5-5% disseminated lenses of pyrrhotite parallel to foliation. In many areas, replacement patches consist of coarser grained plagioclase, quartz, ankerite, chlorite and sulphides. Sulphides are mainly pyrrhotite or pyrite, with local concentrations of one or more of sphalerite, galena and chalcopyrite. Two main alteration assemblages were recognized in trenches. The first consists of semi-massive to massive sulphide dominated by granular pyrite intergrown with abundant dark green chlorite and much less abundant sericite and quartz. The other consists of felsite containing 20-30%, very fine grained, disseminated pyrrhotite. Both types of altered rocks contain anomalous concentrations of base and precious metals.

Frank Creek and SCR project areas

Work by Struik (1983) and Ferri (2001) placed the rocks of these project areas in the Snowshoe Group of the Barkerville terrane. These rocks include, from oldest to youngest, the

Keithley succession, Harvey's Ridge succession and Goose Peak quartzite (Figures 7a and 7b). The Keithley succession consists of micaceous quartz sandstone to siltstone, phyllite and quartz-muscovite schist, in which common pyrite and ankerite porphyroblasts give a brown to rusty brown weathering surface. At the top of the Keithley succession is a distinctive quartzite. Southwest of Browntop Mountain, a section of rusty-weathering quartz-chlorite-muscovite schist of the Keithley succession contains two stratabound intervals of marble 50-75 m thick. This area also contains a quartzite unit that is less than a few metres thick and contains minor disseminated chalcopyrite, sphalerite and galena. The upper contact of the Keithley orthoquartzite is sharp with the dark grey to black siliceous siltstone and phyllite of the Harvey's Ridge succession.

The lower part of the Harvey's Ridge succession is characterized by dark grey to black phyllite, schist, siltstone, siliceous siltstone and sandstone. An important wedge of mafic volcanic rocks occurs west of the Frank Creek fault. The upper contact of the Harvey's Ridge succession is gradational with the overlying Goose Peak quartzite. The contact is placed at the base of the lowermost, thick section of clean, feldspathic sandstone to quartzite. The Harvey's Ridge succession contains abundant interlayers of coarse grained, feldspathic sandstone and wacke, similar to but darker in colour than those of the Goose Peak quartzite.

The Goose Peak quartzite contains micaceous and feldspathic sandstone to quartzite, with minor interlayers of grey to dark grey phyllite/schist, siltstone, wacke and quartz-chlorite-muscovite schist (possibly of volcanic origin). This unit caps many of the higher ridges situated to the east of the Frank Creek fault, including Goose Peak, Badger Peak and Borland Mountain. The Agnes conglomerate occurs in the upper transitional portion of the Harvey's Ridge section and may be a lateral equivalent of the Goose Peak quartzite. The Quesnel Lake gneiss, which intruded the older rocks in the Frank Creek area, was dated at 357.21.0 Ma by the U-Pb zircon method by the BC Geological Survey (Ferri, Hoy and Friedman, 1998).

The stratigraphic section is overturned and warped by a series of broad late-stage (F3) folds. Most rocks are variably slaty, foliated, laminated or schistose. In much of the Frank Creek area, the main foliation (S2) dips moderately to the west or southwest, in contrast to the regional trend of moderate dips to the northeast, such as at the Ace project area. The Frank Creek area is in the nose of a series of broad F3 folds that warp S2 gently to steeply, locally, and plunge gently to the northwest.

Most rocks of the Frank Creek project area are of the Harvey's Ridge succession. At the stratigraphic base of the section (top of Frank Ridge) is a finely banded, tuffaceous, basaltic andesite. Much less abundant are massive, coarser grained flows, one of which contains a few pillow flows, whose orientation supports the overturned model. A transition zone contains intercalations of volcanic and sedimentary rocks and mixed rocks described as tuffaceous phyllite. Stratigraphically overlying this unit (downslope to the north) is a zone of mixed quartz-pebble and sugary quartz sandstone, dark grey to black quartz siltstone and black argillite. Further downslope is a unit at least several tens of metres thick containing abundant angular fragments of quartz and lesser plagioclase ranging 0.5 - 3 mm in size. This unit has been interpreted either as pebbly sandstone to conglomerate or medium to coarse felsic tuff. It contains intervals up to a few metres thick of black, argillaceous schist that is mainly non-graphitic, but contains local zones of significant graphite content.

A broad interval represented by little to no outcrop separates the main zone of quartz-pebble rocks from the stratigraphic level of the massive sulphide. Stratigraphically above the massive sulphide showing is a small outcrop containing abundant zones of pyrite and chalcopyrite replacement and stringer mineralization. The discovery of the massive sulphide zone and its probable genetic relationship to these rocks, which have been interpreted by Payne and some others to be felsic volcanic rocks, indicate that this area has potential for an economic VMS mineral deposit.

Further to the east and northeast is a resistant marker unit of quartzite and locally finely laminated limestone. The interpretation of this unit is not resolved. If the section is overturned, as detailed data suggest, this could be part of the Goose Peak quartzite. Alternately, it might be an interval of the Agnes conglomerate. In some outcrops, a pseudo-conglomeratic texture was produced by deformation. Some of these outcrops contain scattered, rounded boulders up to 25 cm across of quartzite of the same composition as the matrix. If they are actual boulders, the rock would be a true conglomerate. However, the absence of exotic boulder types argues that they are pseudo-boulders and are the products of the regional metamorphism. The limestone is exposed only in the canyon of Frank Creek. The quartzite continues as a significant stratigraphic marker unit to the southwest, where it does not contain any pseudo-conglomeratic textures.

To the east, a major fault (Frank Creek fault) identified in this study, drops the Frank Creek block down with respect to the Goose Peak block to the east, in which surface rocks are dominantly Goose Peak quartzite. Although regionally, F3 folds and L12/L23 lineations in the Goose Peak block also plunge gently to moderately to the northwest, within about 1 km of the Frank Creek fault, the plunge is reversed to gently to the southeast. Further work on the regional folding should be done in order to resolve the facing direction of the stratigraphic section and the direction of movement on the Frank Creek fault. Preliminary work in this study suggests that the stratigraphic section is overturned and the fault is normal.

Cariboo (Maybe) Prospect area

The Cariboo (Maybe) Prospect was explored during 1987 by Gibraltar Mines Ltd. The prospect contains three main stratiform lenses of ankerite-quartz-sphalerite-galena-(pyrite) enclosed in limestone-rich intervals of probable Middle Devonian age (Hoy and Ferri, 1998). These, in turn, are inter-layered with dark grey graphitic phyllite. One prominent quartzite bed occurs a hundred metres stratigraphically below the sulphide-rich beds. The rocks were deformed, sheared and foliated. Some of the lenses probably were formed by late- to post-tectonic remobilization.

Glaciation and glacial deposits

The last glacial stage that affected the Quesnel Highland, the Fraser glaciation, began 30,000 years ago. Much of this ice had melted by 10,000 years ago, but small remnants are preserved high in the alpine areas of the Cariboo Mountains. At lower elevations, glaciers of this age scoured the debris left by preceding ice advances, almost completely destroying them, leaving a chaotic assemblage of unsorted till, moraine and drift, with lenses of gravel and sand that had been roughly sorted by meltwater and rivers, leaving behind beds of silt and clay that were stratified by settlement in ice-dammed lakes. In the Cariboo area, the debris covers bedrock in valleys below 1700 m, leaving typical glacial features such as U-shaped valleys, ice-sculpted drumlins, moraine terraces and glacier and river benches. On the Barker Minerals properties, glacial deposits range from one to a few tens of metres thick. Some glacial till deposits are overlain by well-bedded glaciolacustrine clay and silt deposits up to a few tens of metres thick.

In much of the Cariboo district, a layer of distinctive, hard, compact, semi-rigid blue clay sits either on or slightly above bedrock and acts as "false" bedrock. It was formed from glacial drift left behind by the last ice advance prior to the Fraser glaciation and was compacted by the weight of the Fraser stage ice. In the placer-gold areas of the Cariboo, large amounts of gold were recovered from gravel resting on this clay. In places the clay layer was penetrated by the placer miners to reach richer "pay streaks" on true bedrock below.

Item 10: Deposit Types

The company is exploring the property for the following types of mineral deposits:

Volcanogenic Massive Sulphide (VMS) Deposits

Conformable, semi-massive to massive sulphide deposits of the Besshi type occur in the Kootenay terrane (Goldstream deposit), in the Yukon-Tanana and Nisling terranes and in the Klondike schist. Recent BC Geological Survey work suggests that these may all be part of an elongate terrane that also includes the Barkerville terrane (Hoy and Preto, 1996). Host rocks are deformed complexly and metamorphosed to micaceous quartzite, phyllite and schist, commonly graphitic. Marble and meta-volcanic rock lenses are common. Besshi-type deposits contain pyrite, pyrrhotite, magnetite and chalcopyrite, local sphalerite and rarely galena. Host rocks are mainly sedimentary, commonly siltstone, quartzite and carbonaceous schist near amphibolite (metamorphosed mafic volcanic rocks). The VMS target at the Ace prospect may be of a different type; there, felsic volcanic rocks are present, mafic volcanic rocks are absent or altered so intensely that they cannot be recognized, and galena and sphalerite are important minerals.

Zn-Pb Replacement

The sulphide-rich layers of mineralization of Barker Minerals' Cariboo Zn-Pb deposit have a higher-grade central section up to 1 m. thick of more massive sphalerite with scattered patches of galena, with dispersed and vein sphalerite-galena mineralization extending 2-3 m on either side. The high Zn/Pb ratio, moderate silver content and low gold content are similar to those of many replacement carbonate-hosted Zn-Pb deposits, including those in the Early Cambrian platform carbonates of the Kootenay arc.

Vein and Replacement Gold

Much of the lode and placer gold production from the Wells and Barkerville areas was from-the eastern part of the Barkerville terrane, where most of the important present day lode gold prospects also occur. At the historic mines, the strata trend 315°, dip 45°NE and are cut by north to northeast-trending normal faults dipping 60°E. The zones of economically important quartz veins are contained in graphite-bearing layers near a contact with carbonate-bearing layers. Two sets of quartz veins contain pyrite and gold: transverse veins striking 030° and diagonal veins striking 070°; these contained 15-25% pyrite and 0.3-0.5 oz/st Au. Two other sets of quartz veins do not contain pyrite or gold. Pyrite in altered wall rock contained proportionally less gold. Replacement bodies of auriferous pyrite in carbonate rocks (mainly in the Island Mountain Mine north of Jack of Clubs Lake) accounted for nearly 1/3 of the lode gold mined. Other common sulphides in the mines are arsenopyrite, galena, sphalerite, cosalite (PbBiS), scheelite and bismuthinite (BiS_2). Free gold commonly occurs with nests of cosalite. Gangue minerals are quartz and ankerite/siderite. Wallrock alteration minerals are ankerite and sericite.

Magmatic PGE mineralization

Some of the PGM contents of the Quesnel River placer deposits are probably derived from ultramafic rocks situated along the Eureka fault between the Quesnel terrane and the Barkerville terrane and from mafic and ultramafic rocks in the Quesnel terrane. Barker Minerals' claims in this area were staked because of potentially favorable geology, the recent dramatic increase in the price of PGMs and the apparent lack of previous PGE exploration in this potentially favorable area. Magmatic, dunite-peridotite-hosted PGE mineralization is sought.

Item 11: Mineralization

Providence Mine (Past Producer; BC MinFile 093A 003)

Located in the Blackbear project area of Barker Minerals' property (Figure1, project area #7), the mineralization is comprised of argentiferous (silver-bearing) galena with pyrite, minor sphalerite and gold contained within three subparallel gently dipping quartz veins hosted by metamorphosed tuff, possibly rhyolitic. The main vein is known to be approximately 180 m long and averages 4.5 m in width. A shipment of selected ore from the No. 2 zone assayed 3343 grams of silver per tonne, 45.7% lead, 0.11% zinc and 4.9 grams of gold per tonne (all the above from BC MinFile 093A 003). One grab sample (# 11-07-98-59; Barker Minerals) of outcropping mineralization contained 52% Pb, 0.03% Zn, 142 oz Ag / ton and 0.081oz Au / ton. (Payne, 1999; BC Assessment Report).

Cariboo Zn-Pb Deposit (Developed Prospect; BC MinFile 093A 003) and Foster Showing

Barker Minerals' Cariboo Prospect (Figure 1, project area #2), also known as the Maybe Prospect and classified as a developed prospect, was explored during 1986 and 1988 by Gibraltar Mines Ltd. The prospect contains three main stratiform lenses of ankerite, quartz, sphalerite, galena and minor pyrite enclosed in limestone-rich strata of probable Middle Devonian age. The large Zn/Pb ratio, moderate silver content and low gold content are similar to those of many carbonate-hosted, replacement Zn-Pb deposits, including those in the Early Cambrian platform carbonates of the Kootenay arc. Sampling of the zone intermittently over a 1.6 km strike length returned concentrations up to 15% combined Zn and Pb (Roach, 1997; BC Assessment Report). Limited bulldozer trenching and 21 diamond drill holes by Gibraltar Mines resulted in an estimated historical mineral inventory of 400,000 tonnes with an estimated grade of 4% Zn+Pb (BC MinFile 093A 110). This is not an appropriate category and does not comply with the requirements of current securities legislation. Barker Minerals' grab sample results returned concentrations up to 32.8% Zn, 4.5% Pb, and 63 g/t (2 oz/t) Ag.

Further east and south along the Pleasant Valley Thrust is the Foster mineral showing, situated at the junction of the thrust structure and a north-northwesterly striking fault. Here, abundant finely disseminated pyrite (5-15%) occurs in a zone 30 m wide of unknown strike length in sheared clastic metasediments dipping 40–50 degrees to the northeast The mineralization is accompanied by sericite-quartz-chlorite alteration (Roach, 1996).

Ace Prospect; VMS and auriferous quartz (BC MinFile 093A 142)

Mineralized boulder field:

During 1993, in the Ace Prospect project area (Figure 1, project area #1) at the outlet of a culvert on the "F" spur, a branch of Weldwood's "8400" logging road, a grab sample of sand collected by Louis Doyle (possibly contaminated by another collected 2 km down the road) assayed 129 g/t gold (Lammle, 1996; BC Assessment Report). At the end of October, Doyle staked the Unlikely I and Unlikely II mineral claims on the potential source area. During November of the same year these claims were vended to a newly formed company, Barker Minerals Ltd.

During 1994, the Ace mineral claims were staked around the Unlikely claims on the north-facing slope of Mount Barker. Prospecting, reconnaissance geological mapping and rock sampling, line cutting and soil geochemical sampling conducted during the summer of 1994 helped to locate many glacially transported cobbles and boulders comprised of vein quartz, quartz-pyrite-pyrrhotite and/or semi-massive to massive iron-rich sulphides. (Figures 4, 6a and 6b) Many of the boulders of quartz-rich veins and massive sulphides in the boulder field are sub-rounded to sub-angular. Many of the nearby coarse blocks of country rocks are distinctly angular. These features suggest that the as-yet-undiscovered bedrock source of the boulders is probably close by, possibly beneath glacial till deposits located up-ice (opposite direction of the flow ancient glacial ice) from the present locations of the boulders. The mineralization in some of the boulders of Doyle's boulder field has been compared with that of the Goldstream deposit north of Revelstoke and with that of the Vine prospect in southeastern B.C. (Hoy and Preto, 1996).

The main boulder field, along Weldwood's main "8400" haul road from km 8423 to km 8431, is 8 km long and several hundred metres wide. Many boulders are comprised of quartz veins with minor to moderately abundant amounts of one or more of tourmaline, sphalerite, chalcopyrite, galena and graphite. Grab samples collected from many boulders contained significant concentrations of gold and/or base metals (Figure 6b). The average of 53 widespread float boulders comprised of sulphide-bearing quartz veins was 3.1 g/t Au, with concentrations ranging up to 29 g/t Au. Many of the higher-concentration gold samples contain significant concentrations of lead (1000-2000 ppm), bismuth (100-2500 ppm), selenium (20-50 ppm) and tellurium (10-34 ppm). Several pyrrhotite-rich massive sulphide boulders contain 3-16% Zn+Pb and up to 3 oz/t Ag and Cu stringer material up to 4.1% Cu.

Skupinski (1995) conducted a petrographic examination of Ace boulder samples. "In four polished mounts and two polished sections over 30 grains of native gold were detected. The three largest grains were 0.03-0.05 mm in size. The other 27 grains were 0.01-0.03 mm in size. The grains are subhedral to euhedral in shape, especially those occurring in quartz. Native gold mainly occurs close to the borders between sulphides and quartz. Less frequently, it occurs as inclusions in pyrrhotite and marcasite. Frequently, gold is intergrown with tellurides and native bismuth. As a rule, all above minerals precipitate in close proximity to each other. They tend to occur on the surface of sulphides, or very close to the sulphide edges. Bismuth and telluride grains are always anhedral and up to 0.1 mm in size. By visual evaluation, volumetric contribution of telluride minerals in quartz vein (float boulders) is 100 times higher than that of native gold."

Precious and base metals have been, and continue to be, the major economic focuses of exploration.

Ace area bedrock mineralization:

The widespread presence of 2-5% disseminated pyrrhotite and/or pyrite in felsic volcanic rocks that has been indicated by surface rock sampling and 2 episodes of drilling along the main trend of the Ace zone suggests potential for VMS deposits.

Barker Minerals' trenches (Figure 4a) have exposed zones up to 10 m thick of semi-massive sulphide containing 20-50% pyrite and/or pyrrhotite at its Ace VMS exploration project (Payne, 1998; BC Assessment Report). In the context of the presently accepted VMS models, this unit may represent the footwall of a VMS deposit. Two main alteration and mineralization assemblages were recognized in trenches. The first consists of semi-massive to massive sulphide dominated by granular pyrite intergrown with abundant dark green chlorite and much less abundant sericite and quartz. The other consists of felsite containing 20-30%, very fine-grained, disseminated pyrrhotite. Both types of altered rocks contain anomalous concentrations of base and precious metals.

Diamond drill hole 98-03 (dipping -45°) intersected a stratiform, mineralized layer (Figures 4a and 4a-2). The intersection was 0.75 m long, and contained two massive sulphide intersections, 0.25 m and 0.20 m long, separated by an intersection of calcite 0.30 m long. Diamond drill hole 98-07 (dipping -45°) intersected semi-massive sulphide mineralization for an intercepted length of 0.36 m. The massive sulphide mineralization is composed of massive sulphide minerals of iron, copper and zinc, and carried small amounts of silver and gold. (Table 5c).

Two of the holes drilled during 2002 intersected significant sulphide mineralization near the top of the felsic volcanic rock interval. Ace-02-03 intercepted 3.3 metres of semi-massive to massive sulphide mineralization with anomalous Cu/Pb/Zn metals. Mineralized felsite extended an additional 69 metres down hole below the strongly mineralized layer (Table 5c).

The volume and grade of the mineralization have yet to be determined.

Frank Creek VMS showing (BC MinFile 093A 152)

During 1999, a trench excavated in a zone of weathered massive sulphide boulders near a culvert 2.2 km up the "D" logging road of the Frank Creek project area (Figure 1, project area #4) exposed a stratiform, massive sulphide layer at least 1.2 m thick over a strike-length of 10 m (VMS showing in F-1 target area shown in Figures 5a, 5b, 5c, 8a and 9). The attitude of the layer (strike 167°, dip 32°) is sub-parallel to that of the major foliation (S2) in the surrounding rocks. Additional boulders containing massive sulphide and stringer-style sulphide mineralization were found along strike as far as 150 m away. Reconnaissance VLF and Mag geophysical prospecting indicated the potential presence of a conductor under the D Road. Assay results of grab and chip samples collected from the boulders and outcrop (Table 2) suggest zoned-type, massive sulphide mineralization. During 2000, one chip sample (2306) was collected from a pod of magnetic massive sulphide 25 cm thick located stratigraphically beneath the main massive sulphide layer. Trenching during 2002 exposed small massive sulphide lenses approximately 70m along strike to southeast of the discovery outcrop. Along with the mineralized exposure in TR-BW-10 excavated earlier some 375 metres to the NW of the discovery outcrop, this exposure extends the known strike length of the mineralized zone to 425 metres. The mineralized zone is open in both directions to the north and south.

The volume and grade of the bedrock VMS mineralization have yet to be determined.

Float boulders mineralized with massive and disseminated base metal sulphides (predominantly sulphides of iron) occur locally. Three chip samples, collected from heavily mineralized boulders located in the placer area one km downhill and downstream (F-8 target area), contain anomalous concentrations in base and precious metals, and elevated concentrations of As, Sb, Bi, Hg and Sn. The most probable source for these boulders is from an eroded part of an as-yet-undiscovered massive sulphide body located higher up on the west bank of Frank Creek. Such a body could be the southeast extension of the known massive sulphide body or another separate massive sulphide lens.

Prospecting during the 2002 field season on the F-7 target area resulted in the discovery of massive sulphide float boulders, samples of which contained concentrations of zinc up to 7.3% (Table 2). The F-7 target area has associated airborne and ground HLEM anomalies, and copper, lead and zinc soil anomalies were detected nearby.

Float massive sulphide mineralization has been identified on F-1, F-4, F-7 and F-8 target areas, bedrock massive sulphide mineralization has also been identified on the F-1 and F-4 project areas. Since massive sulphide deposits tend to occur in clusters, the Company's chances for discovery of additional massive sulphide mineralization at other target areas throughout the Frank Creek project area may be enhanced.

Big Gulp showing (BC MinFile 093A 151)

This mineral occurrence located within the F-4 target area of the Frank Creek project area (Figure 1, project area #4; and Figure 8) is comprised of sulphide-rich lenses in metamorphosed, altered, now ankeritic, fine-grained tuffaceous sedimentary rocks of original andesitic basalt composition. Chemical analysis of a composite grab sample collected during 1999 yielded results as 4.7% Zn, 600 ppm Cu, 1,520 ppb Hg, 435 ppb Ag, 205 ppm Cd and 29 ppm Pb (Payne, 2000; BC Assessment Report). A grab sample of this mineralization (#226802) collected by INCO personnel assayed 8.27%Zn and 791 ppm Cu, with traces of Pb and Ag. The volume and grade of the mineralization have yet to be determined.

Sellers Creek showing (BC MinFile 093A 131)

Sulphide minerals occur in the Sellers Creek showing (Figure 1, project area #5; Figures 8, 11a and 11b) as semi-massive and stringer mineralization in float and bedrock of altered intermediate to mafic metamorphosed volcanic rocks (Payne, 2001; BC Assessment Report). Sulphide minerals include pyrite, pyrrhotite, chalcopyrite, sphalerite and galena. Geophysical surveys outlined significant, coincident magnetic and conductive anomalies near the discovery area and an area of Cu, Pb and Zn soil geochemical anomalies. The significant results of Barker Minerals' rock sampling here are given in Table 4 of this report. The volume and grade of the mineralization have yet to be determined.

Comin Throu Bear showing (BC MinFile 093A 148)

At this location within the Black Stuart project area (Figure 1, project area #2), galena and barite (sedimentary or diagenetic mineralization) and quartz veins containing galena, sphalerite and possibly tetrahedrite occur within a dolostone breccia. The volume and grade of the mineralization have yet to be determined. The Company has not sampled this mineralization.

Peacock showing (BC MinFile 093A 133)

The Peacock showing consists of several quartz veins up to about one meter wide in silicified schist. These veins, which strike to the north and dip steeply to the east, are weakly mineralized with galena, sphalerite and pyrite, with associated anomalous concentrations of gold and silver. According to BC government maps (Report of Minister of Mines (1926), page A178), the showing is situated within Barker Minerals' Rollie project area (Figure 1, project area #10). It is thought that the new Besshi-type VMS mineralization described recently in this area by the BC Geological Survey may be related to this old mineral showing, now re-discovered.

The volume and grade of the mineralization have yet to be determined.

Trump showing (BC MinFile 093A 154)

Quartz veining occurs in both argillite and andesite units, but four mineralized quartz veins are hosted by phyllitic argillite. Mineralization consists of discontinuous blebs of argentiferous galena and disseminated pyrite in the quartz veins. Disseminated pyrite also occurs throughout the argillite. Sericitic alteration halos up to 3 meters across are associated with the galena-bearing units. Sample # 98-66 collected by Barker Minerals yielded 56.6 oz/st Ag, 0.55% Bi, 0.019 oz/st Au, and 54% Pb, and Sample # 98-69 yielded 36.5 oz/st Ag, 0.35% Bi, 0.023 oz/st Au and 37.6% Pb (Payne, 1999; BC Assessment Report). The volume and grade of the mineralization have yet to be determined.

Item 12: Exploration

Ace VMS and Vein Gold Project

During 1993, at the outlet of a culvert on the "F" spur, a branch of Weldwood's "8400" logging road (Figure 1, project area #1; and Figure 4a), a grab sample of sand collected by Louis Doyle (possibly contaminated by another collected 2 km down the road) assayed 129 g/t gold (Lammle, 1996; BC Assessment Report). At the end of October, Doyle staked the Unlikely I and Unlikely II mineral claims on the potential source area. During November of the same year these claims were vended to a newly formed company, Barker Minerals Ltd.

During 1994, the Ace mineral claims were staked around the Unlikely claims on the north-facing slope of Mount Barker. Prospecting, reconnaissance geological mapping and rock sampling, line cutting and soil geochemical sampling conducted during the summer of 1994 helped to locate many glacially transported cobbles and boulders comprised of vein quartz, quartz-pyrite-pyrrhotite and/or semi-massive to massive iron-rich sulphides (Lammle, 1996; BC Assessment Report). Many of the boulders of quartz-rich veins and massive sulphides in the boulder field are subrounded to subangular. Many of the nearby coarse blocks of country rocks are distinctly angular. These features suggest that the bedrock source of the boulders is probably close by, possibly beneath glacial till deposits located up-ice (opposite direction of the flow ancient glacial ice) from the present locations of the boulders.

The main boulder field, along Weldwood's main "8400" haul road from km 8423 to km 8431, is 8 km long and several hundred metres wide (Figure 4a and Figures 6a and 6b). Many of these boulders are comprised of quartz vein material with minor to moderately abundant amounts of one or more of tourmaline, sphalerite, chalcopyrite, galena and graphite. Grab samples collected from many boulders contained significant concentrations of gold and/or base metals. The average of 53 widespread float boulders comprised of sulphide-bearing quartz veins was 3.1 g/t Au, with concentrations ranging up to 29 g/t Au (Lammle, 1996; BC Assessment Report) (Figure 6b). Many of the higher-concentration gold samples contain significant concentrations of lead (1000-2000 ppm), bismuth (100-2500 ppm), selenium (20-50 ppm) and Te (10-34 ppm). Several pyrrhotite-rich massive sulphide boulders contain 3-16% Zn+Pb and up to 3 oz/t Ag and stringer boulders up to 4.1% Cu. Precious and base metals have been, and continue to be, the major focuses of exploration.

The 1995 program included prospecting, line cutting, geochemistry, geophysics and geology. This expanded the database and identified geophysical and geochemical targets (Lammle, 1996; BC Assessment Report).

During 1996, Barker Minerals staked 2,590 claim units around the Ace project area and performed limited trenching and geological mapping on the Ace project area (Lammle, 1997; BC Assessment Report). Regional stream sampling and magnetometer surveys were conducted on these claims surveys; this resulted in the discovery of the Big Gulp mineral occurrence of sulphide rich lenses.

A test I.P. Survey was conducted on the central part of Doyle's boulder field by Scott Geophysics of Vancouver, B.C., to improve definition of geophysical targets outlined by the 1995 magnetic and VLF-EM studies (Scott, 1996). Several shallow, discrete targets have been outlined, mainly trending parallel to the regional stratigraphy and mainly indicated represented by coincident resistivity lows, chargeability highs and mag highs (Figure 4a, see resistivity low).

An experimental "E-Scan" 3-dimensional resistivity survey was done by Premier Geophysics of Vancouver, B.C., on part of the Ace (Kloo) grid southeast from the GSC2 fault on lines spaced at 100 m (Shore, 1997). The survey outlines a prominent near-surface resistivity low centred at 10E-36S that extends to depth (Figure 4b).

Results of the 1995 and 1996 GSM-19 magnetometer surveys are shown in Figure 5. The survey defined several, narrow, linear, near-surface anomalies associated with areas of high chargeability. Some magnetic anomalies occur in a circular zone about a magnetic low in the Ace 86 claim in the central part of the property. This low coincides with moderately anomalous values in gold, arsenic and boron in soil samples. As well, this area contains a circular area containing sinuous and irregular VLF-EM high readings.

In 1995 and 1996 on the Ace grid, 4,420 soil samples were collected from the top of the "B" soil horizon, or as close to that horizon as could be determined (Lammle 1996, 1997; BC Assessment Reports). In general, the top of the B horizon was reached at depths between 0.2 and 0.7 m. Figures 4c, 4d and 4e show the results of soil studies for Cu, Pb and Zn. Threshold concentrations used were as follows: Cu - 50 ppm, Zn - 100 ppm, and Pb - 25 ppm. The 1996 results compliment the patterns in the 1995 survey results, and show the more extensive nature of the Pb and Zn anomalies situated along the northern margin of Doyle's boulder field between the GSC-1 and GSC-2 faults.

During October, 1996, preliminary excavator/backhoe trenching in and near Doyle's boulder field was conducted. Thirty-six trenches totaling 260 m were excavated (Figure 4a.) The average depth was 4 m, and a few trenches reached 5.5 m in depth. These exposed bedrock at scattered points within the central part of the boulder field. Acme Analytical Laboratories analyzed 107 grab specimens of host rocks and mineralized zones by ICP, whole-rock and hydride methods. Gold concentrations in rock samples collected from four trenches are anomalous. Two grab samples collected from Trench 30 near the "F" road, contained 1065 ppb and 1386 ppb Au. Grab samples of two siliceous rocks collected from Trench A contained 296 ppb and 77 ppb Au. Two grab samples collected from Trench G contained 213 ppb and 50 ppb Au. Concentrations of gold greater than trace correlate positively with Cu, Pb, Fe, Te and SiO2, and to a lesser degree with As, Bi and Hg. A negative correlation exists between Au and Zn.

During 1997, surface mapping was begun in an area of 2 sq. km in the core of Ace project area. Detailed geological mapping and sampling were conducted in 20 trenches totaling 1084 m in length and in 46 test pits (Payne, 1998; BC Assessment Report). The average depth was 4 m, and a few trenches reached 5.5 m in depth. Anomalous concentrations in rock samples collected from these trenches are shown in Table 1.

The 1997 program shows the continuity over a length of a few km of a belt of metamorphosed, felsic volcanic rocks dominated by plagioclase, and containing 1-5% disseminated pyrite and/or pyrrhotite. In this belt, trenches have exposed zones up to 10 m thick of semi-massive sulphide containing 20-50% pyrite and/or pyrrhotite. In the VMS model, this unit is in the footwall of the massive sulphide deposits. Less abundant units include muscovite-quartz, quartz-muscovite-biotite schist, and slightly graphitic phyllite. In the felsic volcanic unit, two main alteration assemblages were recognized in the field. The first consists of semi-massive to massive sulphide dominated by granular pyrite intergrown with abundant dark green chlorite and much less abundant sericite and quartz. The other consists of felsite containing 20-30% disseminated pyrrhotite. Both types of altered rocks contain anomalous values in base and precious metals.

From the Ace project area, 433 rock samples were collected, of which 343 were analyzed by ICP, whole-rock and Au-assay methods. From the remainder of the property, 151 rock samples were collected, of which 27 were analyzed by ICP, whole-rock and Au-assay methods. Petrographic examinations were made on 49 rock samples collected from the surface of the Ace project area and 7 from various locations in the eastern half of the property.

During the stream-sediment (silt) survey, 201 samples were collected from the Ace project area and analyzed, and 130 samples were collected from the remainder of the property and analyzed. During 1997, 336 soil samples, which had been collected from a grid on part of the Ace project area in 1996, were analyzed. As well, 30 soil profiles at various locations were logged, sampled and analyzed.

At the west end and northeast corners of the Ace grid, 31 km of additional grid lines were cut. The location of the grid cut during previous exploration was partly rectified by surveying junctions of grid lines with roads and with the Little River. This work continued into 1998. On the Ace project area, geophysical surveys included 11.9 km of magnetometer surveying and 8.7 km of beep-mat surveying.

During 1998, geological mapping continued and seven diamond drill holes (Figures 4a, 4a-1 and 4a-2) totaling 1260 m were drilled in the center of the Ace project area (Payne, 1999; BC Assessment Report). The results were used to help identify a VMS environment containing massive sulphide mineralization as indicated by three intersections of massive and semi-massive sulphide mineralization measuring up to 0.36 m at or near the top of a pile of felsic volcanic rocks reaching 81.5 m thick (see Item 13: Drilling, this report). A footwall alteration zone that extends up to a few tens of metres below the massive sulphide zones contains geochemically anomalous concentrations of base metals and precious metals. Forty-five polished thin sections from drill core were examined.

During 2000, 42 km of new grid line were cut and Max-min HLEM surveys (57.5 km) and magnetometer surveys (42 km) were conducted by Peter Walcott and Associates, Vancouver, B. C., resulting in the detection of three new indicated conductors A, B and C (Figure 4a).
The HLEM survey indicated one major east-west trending conductor (Conductor A) 1200 m long and open to the east. Conductivity is poor to moderate. It is associated with a magnetic high on the south side of a broad magnetic low. A one-line, broadly indicated conductor (Conductor B) of poor to moderate conductivity is indicated at the northwest corner of the grid and extends off the grid. It is coincident with the east end of the resistivity low located in the1996 survey of the area to the west. Conductor C is indicated as a weak electro-magnetic anomaly situated near the end of two grid lines 200 m north of indicated Conductor B.

Magnetometer readings were taken over the entire grid at 25 metre intervals on lines spaced 100 m apart. The locations of base lines and several other grid points were determined by GPS. The results of the magnetometer survey correlate well with previous data generated by Barker Minerals in 1996. Broad variations in magnetism trend east-west and probably reflect differences in bedrock lithology, which also trend in that direction. A few local dipoles may represent small bodies of a magnetite-bearing rock such as diorite. Sixteen rock samples were collected from the Ace project area and assayed (Payne, 2001; BC Assessment Report).

Barker Minerals oriented and supported a study of glacial till on the Ace project area being conducted by Peter Bobrowsky (2000) of the British Columbia Geological Survey, with the goal of determining the direction(s) of ancient glacial ice movement in order to assist with locating the bedrock source of mineralized boulders found in previous programs (Figure 6a).

During 2001-2002, the Company conducted limited geophysical survey work (Walcott, 2002) comprising 13.4 km vlf-em, 11.8 km gravity, and 21.3 km mag (Figure 4a) mostly in preparation for the 2002 drilling program, the results of which are described in Item 13 (Drilling), this report. Electromagnetic surveying located previously unknown conductors on the Ace grids. Residual gravity anomalies were obtained over EM conductors on widely spaced profiles, suggesting that mineralization could be a partial causative source of those conductors (Walcott, 2002).

Frank Creek VMS and SCR Projects

During 1996, Barker Minerals staked 2,590 claim units and conducted regional stream sediment sampling and magnetometer surveys; this resulted in the discovery of the Big Gulp mineral occurrence of sulphide lenses (Lammle, 1997; BC Assessment Report).

During 1999, after staking the immediate area of the Frank showing, geological mapping and lithogeochemical sampling were conducted in the centre of the Frank Creek area. When numerous pyrite-rich samples were discovered by Barker Minerals personnel while prospecting above an old trench that contained abundant ferricrete, further exploration of this area was added to the program then underway. The prospecting examination in the near vicinity of the trench was extended. This resulted in the discovery of many boulders of massive sulphide located above a culvert situated 150 m east of the trench along side of the D-Road at 2.2 km up the D road. A trench that was excavated near the culvert exposed a 1.2-m-thick massive sulphide layer over a length of 10 m (Payne, 2000; BC Assessment Report) in the F-1 target area shown in Figures 5a, 5b, 5c and 8a. Significant results of rock sampling are presented in

Table 2 of this report. Additional boulders of massive sulphide and stringer-style sulphide mineralization were found up to 150 m away along strike. Reconnaissance VLF and magnetometer geophysical prospecting indicated the presence of a potential 40 m wide conductor under the D-Road.

The massive sulphide samples collected from the Frank Creek showing and placer boulders found approximately 1.3 km downstream present a wide variation in composition, potentially reflecting the possible presence of a strongly zoned deposit. Such variation is typical of many VMS deposits in the world. Most notable are those samples containing concentrations of Ag greater than 14 oz/t and Pb concentrations ranging between 8 and 15.4%. These also contain Cu concentrations in the 0.3-0.6% range and Zn concentrations in the 1-5% range, and some are anomalous in As (1000-8000 ppm), Sb (300-600 ppm), Bi (55-285 ppm) and Se (50-195 ppm). Samples of the placer boulders are similar but contain much lower concentrations of Ag (2-4 oz/t) and Pb (1.9-4.5%), and containing anomalous Sn (300-500 ppm) and concentrations of Zn up to 5.4% and Ba up to 3% (Formosa 1989). Two massive sulphide samples and two from the stringer zone are rich in Cu (1.7-7.4%); these contain 0.5% Zn and minor Pb and Ag. One grab sample collected from the massive sulphide outcrop by an independent mining company contains 8.2% Zn (Table 2; sample #226808). Many samples carry anomalous concentrations of Au (200-450 ppb).

Several samples contain 10-25% CO_3 as calcite and dolomite; this is probably secondary in mafic rocks, but may be, in part at least, primary in phyllite and shale. Carbonate is not abundant in the felsic lapilli tuff, quartz monzonite, or quartzite. Phosphate is very abundant in two samples that are intermediate between shale and phyllite. The Ti/Zr ratios for quartzite, felsic lapilli tuff and quartz monzonite are low, relatively uniform and similar. Those of phyllite and shale are moderately higher and more variable. Mafic rocks have very high and fairly uniform ratios of Ti/Zr, which are high mainly because of a very high Ti content. Barium and potassium are unusually high in the mafic rocks in comparison with normal mafic rocks; this may be the result of potassic metasomatism.

During 2000, prospecting by Barker Minerals personnel resulted in the discovery of the SCR semi-massive sulphide occurrence (Figures 8a, 11a and 11b) located 4 km west of the Frank Creek Project (Payne, 2001; BC Assessment Report).

Surface geological mapping of an area of about 9 km^2 encompassing the Frank Creek and SCR project areas was conducted by J. G. Payne, Ph . D. The centre of the Frank Creek project area had been mapped during 1999. Geological mapping conducted during 2000 was outwards from the initially mapped area.

Thirty-five rock samples were collected and analyzed from the Frank Creek and SCR properties. The significant results of this work are presented in Table 4 of this report. Petrographic examination was performed on 27 rock samples and 7 massive sulphide samples collected from outcrops, most of which were collected within the Frank Creek project area.

Aggregate lengths of grid line cut in 2000 were: 88 km at Frank Creek Project and 17.9 km at the SCR Project. Magnetometer readings over the entirety of both grids were recorded at 25 m–intervals on lines spaced 100 m apart. On the Frank Creek grid, 39.5 km of HLEM geophysical surveying were completed and 19.9 km were completed on the SCR grid. The HLEM survey was conducted on portions of the grids by using a dipole separation of 100 m on the Frank Creek grid, with local areas of detailed survey conducted at 50-m spacing on lines spaced 100 m apart. The HLEM survey on the SCR project was conducted using a dipole separation of 200 m. The locations of base lines and other key locations were determined by GPS surveying.

During 2000, grids were cut on Frank Creek and SCR areas and ground HLEM and magnetometer surveys were conducted (Walcott, 2000). Previous studies had outlined a coincident VLF-EM and magnetometer airborne anomaly associated with a strong base-metal geochemical anomaly in soil and stream-sediment samples in the area of the massive sulphide outcrop.

The magnetic contrast on the Frank Creek grid is low. Despite this, the magnetic patterns mimic the bedrock geology in some areas. Intermediate to volcanic rocks have moderately higher magnetic susceptibilities than the black argillite and siliceous siltstone unit.

The HLEM survey defined several poor to moderate HLEM indicated conductors in the Frank Creek area (Figure 9). Most are shallow and dip steeply. Indicated conductors A and B trend northwesterly east of the massive sulphide outcrop and are located stratigraphically just above the prominent quartzite-limestone marker that is exposed in Frank Creek canyon. The presence of Cu-rich stringer sulphide outcrops stratigraphically above the massive sulphide outcrop suggests the possibility of stacked massive sulphide deposits and makes the indicated Conductor A an excellent target for exploration for a massive sulphide deposit. Indicated Conductor C is broad, trends northwesterly and appears to dip shallowly to the east. Indicated Conductor D dips steeply. Indicated Conductors E, F, G, J and K are associated with the black argillite-siliceous siltstone unit below the intermediate volcanic rocks. Indicated Conductors A, B, C, D, E, F, G, J and K are associated with intense, coincident Cu-Pb-Zn soil anomalies (Figure 10).

Indicated Conductors H and I are anomalies that occur near the Big Gulp showing (F-4 target area, Figure 9) and are open to the west in an area containing no outcrop. One group of HLEM anomalies is associated with a stratigraphic unit of black argillite and quartz siltstone located structurally just below the volcanic rocks. Another, with a coincident soil base-metal anomaly, is in the altered volcanic rocks slightly higher on the ridge. Some of these anomalies are open at one or both ends. More work will be required to define them more completely.

In the SCR grid area, the magnetic contrast is 3.5 times higher than in the Frank Creek grid area, with a relief of 700 nanoteslas (Figure 11a). A 300-m-wide mag high that trends northwest correlates with the south side of a rib of Quesnel Lake gneiss. A narrower mag high trends in the same direction further to the south. Two moderate HLEM indicated conductors trend across the grid in the same orientation as the magnetic anomalies and one (Conductor A) is coincident with Cu/Pb/Zn soil anomalies (Figure 11b). Indicated Conductor A is narrow, presents poor to good conductivity and is associated with a strong magnetic feature. Indicated Conductor B is a narrow anomaly of poor to moderate conductivity paralleling indicated Conductor A, located to the southwest.

From the SCR area, fifteen rock samples were collected from surface outcrops and boulders of semi-massive sulphide. Anomalous results, as determined by non-statistical consideration of the data, are shown in Table 4, with anomalous concentrations in bold type.

Many of these are silicified intermediate volcanic rocks, some of which contain disseminated to semi-massive sulphides dominated by pyrite. A few samples carried anomalous concentrations of one or more of Cu, Zn and Pb, while a few carried anomalous concentrations of one or more of Co, Bi, Se and Sn. Samples SCR-7 and SCR-7a carried anomalous concentrations of Pb and Zn, but contain only minor concentrations of pyrite.

During the period 2001 - 2002 the Company continued prospecting, geological mapping, geophysical surveys, trenching and bedrock sampling, and began diamond drilling at the Frank Creek VMS project, initially outlining a mineralized zone extending 425 m from the Frank Creek VMS discovery outcrop area northwesterly to Trench BW-10 (Figures 5a and 5b).

Electromagnetic (EM) surveying (Walcott, 2002) extended previously located conductor axes on the Frank Creek (Big Gulp) and Sellers Creek grids. While the Frank Creek VMS discovery outcrop showing was unresponsive to electromagnetic and induced polarization techniques, an area presenting anomalous I.P. chargeability responses was observed just east of the showing (Walcott, 2002). Preliminary gravity profiling on the Frank creek grids over the showing area and previously located EM conductors there failed to show any excess mass associated with them (Walcott, 2002). However, the geophysical contractor expressed concern over the required terrain corrections to the data, given the coarseness of the available terrain data. Re-visiting the data with additional control on the grid areas. Thus, the present data are incomplete, and it may not be appropriate at this point to draw any conclusions regarding apparent lack of gravity anomalies at EM conductor locations traversed.

A series of excavator trenches tested several targets in and adjacent to the F-1 Target Area at the Frank Creek VMS project. Five exploratory trenches were excavated over a total distance of 289 metres up to 70 metres southeast of the discovery outcrop. Trench TR-BW-05 exposed the dark quartz eye phyllite host unit within which were found several small massive sulphide lenses within the projected strike extension of the mineralized zone exposed at the discovery outcrop. TR-BW-04 exposed copper stringer mineralization 60 metres southwest of the F-1 occurrence. Along with the mineralized exposure in TR-BW-10 excavated earlier some 375 metres to the NW of the discovery outcrop, this exposure extends the known strike length of the mineralized zone to 425 metres. The zone is open in both directions to the north and south. The significant assay results of bedrock samples collected from the trenches are presented in Table 2.

During the period 2001-2002 the company conducted prospecting, geological and geophysical exploration work at the SCR area, including 4.2 km Maxmin vlf-em geophysical survey, nine test pits and 35 float and bedrock samples analyzed (Figures 11a and 11b and Table 4).

Cariboo Project

During 1996 Barker Minerals conducted reconnaissance geological mapping and rock sampling, with some grab samples returning results of up to 32.8% Zn, with 2.4% - 4.5 % Pb and 63 g/t (2 oz/t) Ag (Figure 8b, after Roach 1996). The zinc and lead sulphides display enrichment in Hg and Sb, and the zone shows other characteristics of an epithermal system such as overall phyllic and advanced argillic alteration, quartz-carbonate and carbonate stockwork, and crustiform banded massive sphalerite and galena (Roach, 1997; BC Assessment Report).

Quesnel Platinum Project

During 2000, Barker Minerals staked several areas of coincident airborne magnetic anomalies, chromium/nickel/cobalt/gold stream sediment anomalies and mafic to ultra-mafic rocks. The areas staked are near reported occurrences of PGMs in placer concentrates as known from previous unpublished studies and assay results from samples collected from local placer miners during 2000.

Barker Minerals collected and assayed 97 rock samples (Table 7) to confirm the presence of the favourable mafic to ultra-mafic rock types, which are host to PGM deposits in other areas of the world (Payne, 2001; BC Assessment Report). Prospecting by Barker Minerals' personnel resulted in the discovery of a Cu-anomaly in the Geremi Creek area.

Reconnaissance geological mapping and sampling were conducted over portions of these claims in order to identify and confirm the occurrences of mafic and ultramafic rocks. Thin-section petrographic examinations were made on six samples. Geological mapping and rock geochemistry confirmed and characterized the presence of mafic and ultramafic rocks in the property. Several samples collected from hematite-altered amygdaloidal mafic flow rocks are anomalous in Cu, Ag and Hg (Payne, 2001; BC Assessment Report).

During 2000, several reconnaissance ground magnetometer and VLF-EM surveys were conducted by Barker staff along roads, in order to confirm and locate more accurately the anomalies identified in previous airborne surveys. These confirmatory geophysical survey traverses were conducted near the locations of stream samples (Figure 2) that contained anomalous concentrations of PGE's *per* Rublee (1986). Several ground magnetic anomalies confirmed the results of the airborne survey. Several VLF-EM survey traverses show significant crossovers that should be followed up with more detailed geological and geophysical studies controlled by conventional cut grids. In several of the traverse profiles, magnetic anomalies coincide with VLF-EM crossovers.

During the period 2001 - 2002, the Company conducted GPS surveying of claim post locations in order to more accurately define mineral claims in this project area and to obtain required assessment work credits (Walcott, 2001 Assessment Report).

Item 13: Drilling

The Company has conducted two episodes of drilling at its Ace VMS and vein gold project (Payne,1998 and Wild, 2002) and one initial exploratory drilling program at it's Frank Creek VMS project (Wild, 2002)

Ace VMS and vein-gold project

Barker Minerals' initial diamond drilling program at the Ace project (1998: 7 holes, ddh 98-01 to ddh 98-07, totaling 1260 m) targeted felsic strata and geophysical anomalies (chargeability highs, resistivity lows and magnetic highs) located within a zone suspected to be underlain partly by felsic rocks having exploration potential for massive sulphide deposits (Payne, 1999; BC Assessment Report). The holes were drilled between the main trench area and Colleen Road (Figures 4a,4a-1, 4a-2). Summary drilling statistics are shown in Table 5a and Summary Drill Logs are shown in Table 5b. Most of the holes were drilled to the south, dipping at -45°, which is approximately perpendicular to the regional trend of the metamorphic foliation. Two holes were drilled more steeply at -60°and -70° in order to attempt to penetrate specific geophysical anomalies. In some instances, it appears that the holes were not long enough to have reached the intended geophysical targets.

All drill holes of the 1998 program, except 98-05, intersected felsite, whose thickness ranged from 3.5 to 81.5 m. As the holes were drilled perpendicular or nearly so to the regional foliation, all thickness intervals are close to true thickness relative to the metamorphic event (Payne, 1999; BC Assessment Report). Because of the complex folding in the region, it is impossible to estimate true, pre-metamorphic thicknesses. Significant intersections and analytical results are presented in Table 5c. At the top of the main felsic section in DDH 98-03, a mineralized zone was intersected for 0.75 m, containing two intersections of massive sulphide mineralization, 0.25 m and 0.20 m long, separated by an intersection of calcite (probably limestone) 0.30 m thick. The massive sulphide mineralization carries apparently anomalous concentrations of Au, Ag, Cu, Zn, As, Se, Te, Sb, Bi, Mo and Cd, as determined by non-statistical consideration of the analytical data. The footwall alteration zone below the massive sulphide is anomalous in many of the same elements, generally with smaller concentrations than in the massive sulphide. As well, the footwall zone carries anomalous concentrations of Mo. Anomalous metal concentrations in the footwall decrease moderately to rapidly away from the top of the section of felsic rocks. Deeper in the footwall, many samples contain anomalous concentrations of Ag, Mo and Zn. In the centre of the felsite section, hole 98-07 intersected semi-massive sulphide mineralization for 0.36 m, containing anomalous concentrations of Ag, Cu, Bi and Se. Above and below this, the rocks contain anomalous concentrations of Ag, Zn, Mo, Bi and Se. Concentrations larger than 100 ppb Au, 500 ppb Ag, 200 ppm Cu, 70 ppm Pb and 150 ppm Zn are considered geochemically anomalous in this report.

During 2002, two diamond drill holes, ACE-02-01 and 02, were collared in the 16S Zone (Figure 4a-1). The first hole, Ace-02-01, tested a coincident max-min conductor and modest gravity anomaly and is located slightly up-ice from the high-grade (16.4% Zn/Pb) boulders found in previous programs. The top 49.0 metres consist of very siliceous quartz-muscovite schist and argillite. The next 10 metres consist of strongly calcareous marble, calc-schist, and argillite, immediately above a 6.5 metre section of "felsite". The felsite consists of plagioclase or albite with minor micas and quartz. The felsite in this area has concentrations in drill core up to 339 ppm Cu, 568 ppm Pb, and 156 ppm Zn. A narrow intercept (.07 metres) of Au/Bi/Te/W quartz vein assayed 745 ppb gold. Follow up geophysical surveys indicate that the possibility exists that this zone was faulted off where drilling occurred; further interpretation is required to determine the next program for this zone (Wild, 2002).

The second hole, Ace-02-02, tested a magnetic high anomaly but failed to explain the source of this anomaly. This strong, large magnetic anomaly is located up-ice from a zone of Au/Bi/Te/W quartz sulphide boulders that have gold concentrations up to 11 grams per tonne (g/t). It is anticipated that the next phase of drilling will test this high priority target for its economic gold potential. A 7.5 centimetre wide quartz vein at 159.6 metres near the end of hole Ace-02-02 assayed 692 ppb gold (Wild, 2002).

The other three drill holes in the Ace Project area, Ace-02-03 to 05, tested the 5N Zone (Figure 4a-2). All three intersected at least 40 metres of "felsite", which hosts VMS mineralization elsewhere on the property. The fine-grained nature of the rock, its intimate relationship to a series of thin marble bands, and almost regional extent suggest that it may be an exhalative horizon and an excellent target horizon for VMS deposits (Wild, 2002).
Two of the recent holes intersected significant sulphide mineralization near the top of the interval. Ace-02-03 intercepted 3.3 metres of semi-massive to massive sulphide mineralization with anomalous Cu/Pb/Zn metals. Mineralized felsite extended an additional 69 metres down hole below the strongly mineralized layer.

In drill hole Ace-02-04, a ten metre interval of mineralized felsite between 43.8 - 53.8 metres, was highly anomalous for base and precious metal with concentrations up to 663 ppm Cu, 855 ppm Zn, 704 ppm Pb and 575 ppb Au. This hole was collared within 400 metres of a cluster of previously identified high grade quartz vein float boulders assaying up to 29 g/p/t Au, and zinc-lead mineralization (10% Zn, 2%Pb) in bedrock.

The last hole, Ace-02-05, was collared on L5N at 6+25W and drilled vertically to test a coincident ground magnetic high and subtle gravity anomaly. It was collared in the felsite unit, likely below the sulphide horizon (Wild,2002).

Felsic volcanic rocks encountered during this drilling program consist mainly of extremely fine-grained plagioclase with minor biotite and/or muscovite. Many contain minor to abundant replacement and recrystallized patches of coarser grained plagioclase, with or without minor to abundant quartz, ankerite, muscovite and pyrite. Some surface samples, previously described as metamorphosed diorite, were reinterpreted as recrystallized and replaced felsic volcanic rocks. Some felsic volcanic rocks contain zones up to several metres wide of weak to strong biotitic alteration, some of which occur in broad, diffuse envelopes about quartz-sulphide veins. In these zones, pyrrhotite is replaced by coarser-grained porphyroblasts and lenses of pyrite.

Veins containing anomalous metal concentrations are of three types. Quartz-pyrrhotite veins in DDH 98-01 carry anomalous concentrations of Cu, Ag and Bi. Quartz-pyrrhotite-tourmaline veins in DDH 98-02 carry anomalous concentrations of Au, Ag, Cu, Zn, Bi, Se and Te. A quartz-pyrrhotite vein in DDH 98-07 carries anomalous concentrations of Au, Ag, Cu, Pb, Bi and Se. Quartz (+pyrrhotite) veins intersected in DDH02-01, DDH 02-02,DDH 02-03 and DDH 02-04 carry anomalous metal concentrations of Au, Ag, As, Sb, Bi, Te, W, Mo and Tl with concentrations of gold up to 764 ppb over 7 cm in DDH 02-01, 692 ppb over 7.5 cm in DDH 02-02 and 526 ppb over 25 cm in DDH 02-04. All six samples from DDH02-01 had geochemical patterns anomalous in Au, Ag, As, Sb, Bi, Te, W, Mo and Tl. A zone from 42.5 metres to 58.6 metres in DDH02-04 was anomalous in Au, Ag, As, Sb, Bi, Te, W, Mo and Tl. Numerous barren quartz and barren quartz-chlorite veins are present in drill core in many rock types and are particularly abundant in some intervals of quartzite and quartz-rich schist.

Comparison of drill-core data with previous analyses of boulders indicates that the massive sulphides and auriferous quartz-sulphide veins in the boulders and corresponding materials in the drill cores may have come from the same environment. This and the general angular appearance of the boulders and other large blocks of bedrock with which they occur in the overburden suggest strongly that the boulders are from a nearby bedrock source, probably slightly up-ice from their present locations.

Frank Creek VMS Project

During 2002, the Company conducted an initial exploratory diamond drilling program (7 ddh, 813m in total) in the vicinity of the Frank Creek massive sulphide discovery outcrop and nearby geological and geophysical targets (Table 6a;Table 6b). All of the initial drill holes were located within the F-1 Target Area (Figures 5a, 5b and 8a).

The drill core obtained during this initial exploratory program contains intervals of Cu-Zn-Pb (+/- Au, Ag) massive sulphide mineralization that are significant examples of ore formation processes having occurred on the property.

The mineralizations encountered in the core (Table 6c) are similar to that exposed at the discovery outcrop where a massive sulphide layer is exposed for some 3.5m length and 1.5 metre width (Lane 2000; Hunter Dickinson, 2000; and Hudson Bay Exploration & Development

2001). There, concentrations of metals from grab samples of the outcrop massive sulphides ranged up to 4.4% Cu, 8.2% Zn, 1.1% Pb, 14.8oz/t Ag and 854 ppb Au. A chip bedrock sample across .77 metres exposed width assayed 2.1% Cu, .34% Zn, .11% Pb and 69 ppm Ag.

This Besshi-type VMS polymetallic mineralization occurs in drill core in significant intervals (up to 0.4 metres) and contains significant concentrations (up to 3.4% Zn, 2.1% Cu, 2.8 oz/t Ag, 0.53% Pb and 746 ppb Au) within larger weakly mineralized units up to 52 metres wide in drill core intercepts that contain widespread disseminations of these metals.

Four of the 6 holes tested the Frank Creek mineralized horizon, the other two tested the immediate structural footwall, to a wide, conductive graphitic fault zone. FC-02-01, 05, and 06 intersected disseminated to semi-massive sulphides, dominantly pyritic, near the projected location of the mineralized horizon, downslope (northwest) of the original Frank Creek (F-1) Showing. FC-02-03 tested under the showing and intersected approximately 50cm of pyritic massive sulphide similar to that in the showing, caught in a large gougy fault zone. FC-02-02 and 04 tested strong max-min conductors and intersected significant graphitic shear zones in carbonate-altered metasedimentary rocks.

Item 14. Sampling Method and Approach

Rock samples

Reconnaissance rock samples were all grab samples. Outcrops and boulders containing sulphide mineralization were sampled by grab or chip sampling. All exposures of bedrock in trenched areas that contain sulphide mineralization were channel sampled. All drill core containing sulphide mineralization associated with the VMS targets was sawn in half with a diamond saw. One half of each sample interval was sent out for assay, while the other half was saved for future reference.

Geochemical soil and stream-sediment samples

Soil samples were collected with grub hoe and trowel, from near the top of the "B" soil horizon, or as near to that horizon as could be reasonably determined under field circumstances. Samples were collected and placed in standard kraft paper envelopes manufactured for the purpose. The sample bags were labeled with the station and line. On return to base camp in Likely, the samples were air-dried in their bags prior to shipment to the lab. At the lab the samples were oven-dried, then sieved to -80 mesh and aliquots taken for testing.

Stream-sediment samples were collected by digging silt from the active steam bed, screening out coarse material in the field and sending the wet sample to the lab, where the -80 mesh fraction was obtained and analyzed.

Factors to affect accuracy and reliability of samples

A few samples of drill core were from intervals for which drill recovery was significantly less than 100%. These include some of the intervals of massive sulphide.

Sample quality and representation

Samples are of high quality and are representative of the population from which they were collected.

Geological controls

Sampling of drill core from the Ace drilling programs was mainly from distinctive felsic volcanic rocks and massive sulphide, with a few samples from interlayered limestone near the stratigraphic top of the section. As the zone of mineralization was up to 80 m wide, sampling was done mainly at 1-metre intervals, except where significant geological contacts indicated that sample borders should be at such contacts, e.g., of the massive sulphide.

Sampling of drill core from the Frank Creek VMS initial drilling program was mainly from distinctive sediments and metasediments. A total of 300 samples were collected from the core by sawing marked intervals, bagging half the core from the interval and retaining the other half as a permanent core record. A total of 178 samples were collected from the Ace core and 112 samples from Frank Creek core. Samples and check samples were analyzed by Eco-Tech Laboratories (Kamloops, BC), Acme Laboratories (Vancouver, BC) and Chemex (Vancouver, BC) by standard ICP analysis, with Au, Pt, and Pd measured by fire assay. Cu, Pb, Zn samples above 10,000 ppm were analyzed by assay method. A suite of samples from a number of holes in both the Ace and Frank Creek project areas have also been analyzed for whole rock major elements, trace elements and rare earth elements in order to better characterize host rocks.

Significant sample results are shown in various tables comprising this report.

Item 15. Sample Preparation, Analyses and Security

Core samples were cut and bagged at Likely under the supervision of John Payne, Ph. D., or Christopher J. Wild, P.Eng. Subsequently, the company re-sampled the core in order to obtain assay data on discreet features and intensely mineralized sub-intervals comprising the mineralized zones, and to facilitate additional whole rock chemical analyses. The samples were stored in a locked garage until shipment. Contract companies specializing in this type of work collected the geochemical exploration soil and stream sediment samples. Barker staff or contract geologists collected rock samples.

Barker Minerals Ltd. personnel shipped samples to the analytical laboratories.

Pulps, rejects and rock reference examples are stored at Barker Minerals' facilities in Langley. The drill core from the 1998 and 2002 drill programs is stored at the company's field camp in Likely, BC.

All assay and chemical analytical work was performed by certified laboratories. Acme Analytical Laboratories of Vancouver, B. C, performed most of the soil, stream and rock assays. Some check assaying was completed by Activation Laboratories in Ontario and Chemex of Vancouver. The results compared well.

Fire assay

Traditional 30g lead fire assay followed by acid dissolution of the bead and ICP analysis of the resultant solution was used to determine gold, platinum, palladium in rock samples.

ICP

Normally Acme Laboratories provides a 30-element inductively-coupled plasma (ICP) analysis for soil and rock samples. The ICP method involves dissolving 0.5 grams of -80 mesh sample in aqua regia (1hr @ 95°C), and the resulting solution is then aspirated into an ICP emission spectrograph, which determines the concentrations of the 30 elements. A standard "wet" geo-analytical method involving dissolution in aqua regia, extraction by MIBK and determination by graphite furnace atomic absorption spectroscopy was used to determine trace geochemical concentrations of gold in soil samples.

Ultra-trace ICP

Sample preparation and dissolution are the same as for standard ICP analyses. However, much larger splits are used, usually 10 grams. This allows analysis of a broader range of elements with much greater sensitivity and accuracy, using the same ICP emission spectrograph.

Hydride Method

Sample preparation and dissolution are the same as for ICP with 0.5 gram splits. As, Bi, Sb, Se, Te and Ge are then precipitated in a 50% ammonium hydroxide solution. After centrifuging, the precipitate is separated and re-dissolved in aqua regia. The hydrides of the analytes are then volatilized with borohydride solution, and determinations are made via the ICP emission spectrograph.

Whole rock analysis

Samples are dried and crushed to -80 mesh. Then a 0.2 gram sub-sample is placed in a crucible with $LiBO_2$ flux, and melted during 25 minutes at 1025°C. The molten material is then dissolved in nitric acid, and the resulting solution is then aspirated into the ICP emission spectrograph for determination of the concentrations of major oxides and elements.

High-grade gold samples were analyzed by a second lab as a confirmation of the high-grade gold bearing nature of the samples (Table 8). Selected high-grade gold samples were examined by the Geological Association of Canada (Harris 1995), Dr. Payne (1994), and Dr. Skupinski (1994-5) whose petrographic reports confirmed the presence of gold and its associated minerals.

Occasionally, when high-grade gold results were obtained, Acme normally ran a repeat analysis of a sub-sample of the pulp of the sample.

Barker occasionally submitted duplicate samples in order to check the precision of the analytical work being performed. In these instances, a rock sample would be divided into 3 similar portions, with 2 going for analyses with different sample numbers to the same or different laboratories and the 3rd sample being stored for future reference or confirmation purposes.

The sampling, sample preparation, security and analytical procedures are adequate for the present stage of exploration of the property.

Item 16. Data Verification

The author has verified locations of mineralization and important samples collected on the property or has relied on written reports of other professional geoscientists and engineers associated with the project from time to time. The Company has preserved the Certificates of Assay as issued by the accredited geo-analytical laboratories per each sample and test.

The analytical laboratories used standard reference materials and calibration solutions, repeat measurements on solutions and re-assays of pulps in order to insure quality control and data verification in regard to chemical analyses. Occasional cross-check analyses were performed by other certified laboratories, and the results compared favorably (Table 8).

Certificates of Assay were issued in regard to all chemical analytical data and these are appended to the various Assessment Reports pertaining to Barker Minerals' exploration of the property as filed with the Mineral Titles Branch of the Government of the Province of British Columbia. These reports are listed in the references section of this report.

The geophysical surveys were conducted to industry standards by contracted professional geophysicists, including concurrent in-field checks on the validity of the data. Barker Minerals' staff have occasionally verified geophysical data *via* independent traverses with its own equipment.

Numerous samples were collected and analyzed independently by the Geological Survey of Canada (GSC), the British Columbia Geological Survey (BCGS) and several major mining companies as results of property reviews and visits, and the analytical results obtained independently compare well to Barker Minerals' results for similar material collected from the same locales.

Item 21: Interpretation and Conclusions

General

The Barker Minerals VMS and Vein Gold exploration Projects have been visited and examined, in various degrees of detail, by numerous independent professional geo-scientists and engineers, some associated with major mining companies or the BC Geological Survey. They have relayed to Barker their views on the exploration potential of the property and Barker Minerals' management and performance. The following are verbatim comments that this author includes in this Report in order to more fully appraise the reader of other professionals' views on Barker's property and Projects.

Charles R. Lammle, P. Eng. (1995):
"Realizing that he had made a good discovery, and that he was dealing with mineralization of complex mineralogy, he opened a line of communication for advice and support regarding the Ace property with senior provincial and federal government geologists. These geologists were solidly supportive, and results from the property continued to be encouraging."

Strathcona Mineral Services Ltd. (1998):
"Barker Minerals has assembled an impressive amount of information on the Ace and surrounding claims"

V. Preto, Ph. D. , P. Eng., BC Geological Survey (1995):
"Congratulations on the report and all the work that has been done - both excellent"

T. Hoy and V. Preto, Geologists, BC Geological Survey (1996):
"Three deposit types were observed on the tour: Au-quartz veins, carbonite-hosted Zn-Pb and besshi-style massive sulphide. We consider the massive sulphide potential the highest, mainly because of the size and regularity of both geochemical and geophysical anomailes and the amount of massive sulphide float."

Bob Lane, Geologist, BC Geological Survey (1996):
"I am especially interested in the Ace property, which I view as a significant new discovery-one that has ramifications for exploration in Downey succession rocks.

Bob Lane, Regional Geologist, Ministry of Energy and Mines
R. Lane and K. McDonald, Geologist, BC Geological Survey (2000):
"The Ace's host rock lithology and metal content suggest an affinity to well explored gold quartz veins of the Yanks Peak and Cow Mountain areas to the north. The geological setting, style of mineralization and geochemistry suggest an analogy to the "Plutonic-associated" or "Pogo-Type" Gold vein model."

British Columbia Mineral Exploration
Review 2000
Information Circular 2001-1
Ministry of Energy and Mines (Tom Schroeter)
"In the Likely area, 75 kilometres northeast of Williams Lake, Barker Minerals Ltd. continued geological mapping and geophysical surveys over its Frank Creek and Ace volcanogenic massive sulphide showings. Although the felsic rocks at Frank Creek are different from those at the Ace showing to the northeast, the proximity of the two zones in the Barkerville Terrane enhances the potential of discovery of more VMS deposits within the belt".

R. MacDonald, Teck Exploration Ltd. (1999):
"High grade assays from the Frank Creek property visits and Barker Minerals' exploration results from the Ace and other prospects underscore the significant untapped exploration potential of the Likely area. Furthermore, we are encouraged by the variety and widespread distribution of base and precious metal mineralization over the entire property package."

Robert C. Bell, INCO Technical Services (2000)
"We appreciate the excellent cooperation that we have received from you and your team in aiding us in our review, and commend you for the highly professional manner in which you have carried out your exploration and compiled the resulting data."

Ace VMS and Vein Gold Project

The Ace property is underlain mainly by a section of metamorphosed sedimentary and volcanic rocks. The most important sedimentary types are quartz-muscovite schist, quartz-chlorite-muscovite schist, quartzite, quartz-rich schist and quartz-muscovite-biotite schist. In the eastern half of the property, garnet is common, especially in muscovite-rich and biotite-rich schist. Minor argillite, calcareous argillite and limestone are most common directly above the massive sulphide lenses. Deeper in the section on Barker Mountain and along Ishkloo Creek are minor amphibolite and biotite amphibolite.

A volcanogenic massive-sulphide environment is associated with metamorphosed felsic volcanic rocks along the trend of Doyle's boulder field, where many boulders containing gold-bearing quartz veins and gold-bearing massive sulphides have been found. Drill holes intersected a zone of metamorphosed felsic volcanic rocks up to several km long and 80 m thick. These felsic rocks are dominated by plagioclase with minor to moderately abundant quartz, muscovite and biotite.

According to very recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), the host rocks of the Ace mineralization are relatively metamorphosed and structurally deformed. They include a variety of quartz-muscovite-chlorite biotite garnet schists, a pale plagioclase -bearing rock up to several tens of metres thick that has been previously referred to as "felsite", and minor argillite and marble. Lithogeochemical data indicate that the "felsite" in fact is compositionally similar to calc-alkaline andesite. However, due t o metamorphism and recrystallization, it is not certain if it represents a true volcanic rock or a clastic sediment such as greywacke. Amphibolites which probably represent mafic volcanic rocks are also present within 1-2 km of the Ace sulphide occurrence.

At the top of the main felsic section in DDH 98-03 is a zone 0.75 m thick, containing two intersections of massive sulphide mineralization 0.20 m thick and 0.25 m thick, separated by an intersection of calcite 0.30 m thick. The massive sulphide mineralization carries anomalous concentrations of Au, Ag, Cu, Zn, Bi, Se and Mn, as determined by non-statistical consideration of the analytical data.

A few massive sulphide samples collected from trenches in this zone contain up to 2% Zn. A similar zone of semi-massive sulphides 0.35 m thick occurs in the centre of the section of felsic volcanic rocks in DDH 98-07. Massive sulphide boulders containing large concentrations of Pb and Zn occur in the Colleen Road area, but their bedrock source has not been found.

In some strongly altered footwall rocks, the host rock was replaced completely by massive, fine-grained pyrite and dark green chlorite. Others contain 20-30% disseminated, very fine-grained pyrrhotite. Many altered felsic rocks contain anomalous concentrations of base metals and precious metals, in part associated with recrystallized and/or replaced patches dominated by one or more of plagioclase, quartz, ankerite and sulphides. Anomalous concentrations increase in footwall rocks near the stratigraphic top of the main felsic section. These patterns show characteristics of footwall rocks beneath a typical VMS deposit.

The drill cores contain a few important faults up to a few metres wide. Because of the wide spacing of the drill holes, these could not be correlated between holes. The moderate to high graphite content of many of these makes them potentially significant conductors.

Most of the geophysical anomalies obtained in earlier studies have yet to be tested or explained. The main geophysical and geochemical anomaly at the western end of the main trench area is open to the west in an area that is interpreted to be underlain by felsic volcanic rocks. This extends west of the massive sulphide occurrence in DDH 98-03. Vectors in the thickness of massive sulphides, intensity of alteration and geochemical anomalies and thickness of the felsic volcanic section indicate that the area west of DDH 98-03 may contain an important exploration target. Another exploration target is indicated by a broad geophysical resistivity anomaly situated northeast of the area of drilling (Figure 4a). On surface, this area contains rubble of felsic volcanic rocks and abundant boulders of quartz veins containing anomalous concentrations of base and precious metals. A third important target is the elongate HLEM anomaly in the southeastern part of the project area (Conductor A, Figure 4a).

Delineation of the "felsite" unit lead to the discovery of massive to semi-massive mineralization in the 5N Zone, an area that was extensively trenched prior to 1998. Three holes drilled into the zone in 2002 have defined the felsite package as a unit some 60-80 metres thick, possibly thickened by a series of tight to isoclinal folds. Immediately to the west in ACE-02-05 and the east in DDH-98-04, the unit breaks up into a series felsite layers, separated by weakly chloritic quartz-muscovite schists. In the 16S Area, ACE-02-01 was collared to test coincident max-min conductor and subtle gravity anomaly on line 16S. Wild indicated that both the conductor and gravity were caused by a 6.5-metre thickness of felsite with up to 10% pyrite and pyrrhotite, however follow-up geophysical surveys suggest that this hole may not have tested the intended geophysical anomalies, as the zone may be faulted off. The follow-up geophysics defined potentially significant targets situated to the west and also possible faulted off extensions of the favorable horizon to the east (Walcott, 2002) . Drill hole ACE-02-02, was collared immediately above the felsite and tested the footwall near a large ground magnetic high to the south. That anomaly was not adequately explained in the core. (Wild, 2002)

The felsite unit is reportedly made up of up to 80% plagioclase (Payne, 1998), although at least one sample was composed of mainly albite (Walus, 1997). The fine-grained nature of the rock, its intimate relationship to a series of thin marble bands, and almost regional extent suggests that it may be an exhalative horizon and an excellent target horizon for VMS deposits. Sulphide mineralization in 98-3, ACE-02-03, and ACE-02-04 is located near the structural top of the unit.

Gold-bearing quartz veins in this area occur in two main modes. The first are as early veins that were deformed strongly with the enclosing host rock. The second are as late crosscutting veins, many of which trend northeasterly. Many of the quartz veins occur in the same general area as the felsic rocks and boulders of massive sulphide.

The Ace project area has significant similarities to the Intrusion Related Gold System model setting as advanced by Lang, Baker, Hart and Mortenson (Society of Economic Geologists Newsletter #40, January,2000), as it contains geology and mineralogy similar to that of the Yukon and Alaskan settings that host a variety of intrusion related gold deposits (Lane, 2000) such as Fort Knox (158 Mt@ 0.83 g/t Au; Bakke, 1995), Brewery Creek (13.3 Mt @1.44 g/t Au; Diment and Craig, 1998), Dublin Gulch (50.3 Mt @ 0,93 g/t Au; Northern Miner, 1997), True North (16.8 Mt @ 2.5 g/t Au; Harris and Gorton). The newest and highest-grade discovery in the belt is the Pogo deposit (Smith et al., 1999; 9.98 Mt @ 0.52 oz/t Au). According to independent geologists of Teck Exploration Ltd., who expressed that they wish to continue close monitoring of the Barker projects, it is clear from the data that multiple mineralizing episodes are present throughout the property (pers. comm., R. Macdonald, Teck Exploration Ltd., December 1999).

Frank Creek VMS Project

The Frank Creek area contains an important massive sulphide occurrence situated near the stratigraphic top of a fragmental, felsic volcanic rocks or feldspathic arkose. This overlies, in order, a section of black argillite and siltstone and an intermediate to mafic volcanic sequence of flows and fine fragmental rocks. Associated with the massive sulphide zone is a Cu-rich zone of stringer and replacement mineralization. The stratigraphic section is on the overturned limb of one of a set of major, southwesterly verging, F2 folds. All the rocks are metamorphosed, which has inhibited interpretation. The discovery of pillow structures in mafic volcanic rocks in Frank Creek indicates a sea-floor subaqueous environment, thereby enhancing the potential for further discoveries of massive sulphide deposits in this belt of rocks. The association of the Frank Creek massive sulphide and the Ace massive and stringer sulphide mineralization to volcanic rocks enhances the potential for discovery of additional VMS mineralization not only the Goose Range, but throughout the entire Barkerville terrane.

According to recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), original stratigraphic relations and rock textures are much better preserved here than at the Ace VMS environment. Host rocks include quartz-eye-bearing sandstones, siltstones, shales, and minor mafic sills. Work is in progress to determine if the quartz-eye-bearing sandstones represent volcanic crystal tuffs or clastic sediments. Pillow basalts are present 1-2 km south of the Frank Creek sulphide occurrence and basaltic volcaniclastic beds also outcrop in the area, although their stratigraphic position relative to the mineralization is currently uncertain. Nonetheless, the presence of extrusive rocks raises the possibility that mafic volcanism was occurring near the time of Frank Creek mineralization.

The massive sulphide outcrop, spatially related boulders and the Cu-rich stringer and replacement zone contain anomalous to potentially economic concentrations of Cu, Pb, Zn and Ag. Many samples are anomalous in Au and some contain anomalous concentrations of As, Bi, Cd, Hg, Sb, Tl and Mo. Concentrations of As, Bi, Hg, Mo and Sb also are anomalous in stream-sediment samples near other airborne geophysical anomalies. The HLEM survey suggests the presence of another massive sulphide target stratigraphically a few tens of metres above the massive sulphide outcrop. During 2001-2002, trenching and drilling at the Frank Creek VMS project were largely successful in exposing the mineralized zone over a strike length of approximately 425 metres. Based on the results to date, further trenching and drilling are recommended (Item 22, this report).

According to recent preliminary work (pers. com., Dr. Tim Barrett, Ore Systems Consulting, September 26, 2002), the Ace and Frank Creek sulphides occur within the late Precambrian to lower Paleozoic Snowshoe Group of the Barkerville terrane. Based on regional geology, the paleotectonic setting of the Snowshoe Group is generally considered to have been a marine continental shelf which accumulated sandy to muddy clastic sediments and locally mafic to andesitic volcanic rocks. The Cu-Zn-Pb-bearing massive sulphides at Ace and Frank Creek are at least partly hosted by sedimentary rocks. Relative to typical VMS deposits, the massive sulphides are locally enriched in trace metals such as Sb, As, Bi, Hg, In and Sn, which is consistent with derivation of at least some of the metals from hydrothermal leaching of an underlying sedimentary sequence. Possible models for the setting of the massive sulphides, at this early stage of exploration, range from a sediment-dominated continental margin with local, mainly mafic volcanism along extensional faults, to a deeper-water Besshi-type setting, to a sediment-covered spreading axis. The proposed stratigraphic and lithogeochemical studies (as recommended also by this author, Item 22, this report) should serve to define the paleotectonic setting of the mineralization, the most favourable host lithologies, and the distribution of hydrothermal alteration, which in combination will provide a more focused exploration model for the area.

According to Wild (2002), former Chief Geologist of Goldstream Mine, near Revelstoke, B. C., the geological setting, mineralization and host rocks are all remarkably similar to the Goldstream Mine mineral deposit, which produced more than 2 million tonnes of ore at a grade in excess of 4.0% copper and 2.2% zinc.

The Big Gulp showing (F-4 target area) contains anomalous concentrations of Zn in strongly altered intermediate to mafic volcaniclastic rocks associated with a series of east-west-trending HLEM anomalies (F-4 in Figures 8a and 9). This may be indicative of a stringer zone associated with a massive sulphide deposit.

SCR Project

A significant, new VMS prospect has been identified with the discovery of semi-massive sulphide and stringer sulphide mineralization in float and bedrock of altered intermediate to mafic volcanic rocks of the Sellers Creek Road area. Sulphide minerals include pyrite, pyrrhotite, chalcopyrite, sphalerite and galena. The geophysical surveys successfully outlined significant, coincident magnetic and conductive anomalies (HLEM) near the discovery area and near the area of Cu, Pb and Zn soil anomalies. The presence of volcanogenic massive sulphides at the Ace, Frank Creek and SCR properties shows that potential exists for VMS deposits across the entire width of the Barkerville terrane (Payne, 2001; BC Assessment Report).

Quesnel Platinum Project

Reconnaissance geology and thin section petrology confirmed the occurrences and character of mafic and ultramafic rocks. These are probable sources of some of the PGMs in the placers of the Quesnel River and its tributaries. Several samples from hematite-altered amygdaloidal mafic flow rocks are anomalous in Cu, Ag and Hg. A recent BC Geological Survey study confirmed the presence of PGM's in the Quesnel River and some of its tributaries draining Barker's Quesnel Platinum Project areas, especially Black Bear Creek (Levson *et al*, 2002)

Item 22: Recommendations and Proposed Exploration Budgets

Various geological and mining professionals have been involved in the Barker Minerals Projects or have reviewed the Company's exploration work and made field examinations of the active projects. The present author has consolidated the recommendations with which he is in agreement, has added his own recommendations and has added emphasis, in some cases, to the recommendations of others.

Ace Volcanogenic Massive Sulphide and Vein Gold Project

Geological mapping should continue in order to improve understanding of the regional structure and the local geology of areas of felsic volcanic rocks that have not yet been examined. This additional mapping should be integrated with that being done between the Ace and Frank Creek areas by Ferri and others of the B.C. Geological Survey. Wild (2002) recommended extensive whole rock analysis to better characterize the felsite unit. Independent geological consultants of Strathcona Mineral Services (Toronto, Canada) toured the Ace Project and after inspecting core from the drill programs of 1998 and 2002 recommended further work including delineation of the felsite unit through mapping, soil geochemistry and geophysical surveys, followed by trenching where possible and drilling of targets which are selected by the combination of magnetic, MaxMin and gravity geophysical surveys. They are in agreement with Wild (2002) and this author that an effort should be made to determine the origin of the felsite, as this has a bearing on the style of massive sulphide deposits that may exist in relation to this unit on the Company's property.

Barker Minerals has very recently enlisted the services of Dr. Tim Barrett and Dr.Wallace MacLean of Ore Systems Consulting to determine the stratigraphic and alteration relations of the rock sequences hosting the massive sulphide mineralization on the Ace and Frank Creek properties using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography. Ore Systems Consulting has previously worked on numerous volcanic-associated massive sulphide deposits, including those in the Noranda and Matagami camps of Quebec, the Timmins area of Ontario, the Cordillera of British Columbia, Sweden, Wales, Portugal and the Philippines, and have published numerous research papers on these areas, thus this author recommends that they determine the paleotectonic and paleoseafloor setting of the mineralization in each area, as this will have a direct bearing on the VMS model used during exploration. This author also recommends work to determine the chemical identification of the main mineral host rock units at the Ace and Frank Creek Projects in order to trace these favorable units laterally during exploration. The quantitative determination of the degree of alteration is also recommended in order to determine the most intense portions of the hydrothermal systems on each property, as those areas containing a combination of favorable stratigraphy and strong hydrothermal alteration have the greatest potential for economic VMS mineralization. The results should be combined with known geophysical and soil-geochemistry anomalies established both by earlier surveys and potential 2003 surveys in order to help select high-priority drilling targets.

A study should be undertaken to better understand the genesis of the high grade gold/bismuth/tungsten/telluride mineralization present in the very numerous float boulders found on the property. This will assist in developing specific strategies for exploration for the bedrock source(s) of these boulders and similar occurrences.

The HLEM and geochemical surveys should be extended to the west and to the east of the present surveys along the trend of the felsic volcanic rocks. Similar surveys should be conducted over the belt of felsic volcanic rocks north of Little River. The main HLEM electromagnetic anomaly discovered during 2000 has been tested by two gravity profiles which identified excess mass on both test lines that was co-incident with previously defined HLEM (Maxmin) anomalies. Similar gravity profiles should be run across the core of the altered felsic volcanic rocks defined in the 1998 drill program.

An IP survey should be conducted in order to verify the low resistivity e-scan-indicated target.

Additional detailed ground magnetic surveys are recommended in order to continue to identify drill targets which could be possible bedrock sources of the numerous clusters of magnetic pyrrhotite, gold-rich quartz sulphide boulders occurring on the Ace property.

Trenching should be conducted on anomalous zones where the thickness of the glacial overburden allows.

Diamond drilling should test targets identified by the existing geological, geophysical, geochemical and trenching results and by additional work that will be conducted in the first part of the 2003 field season. Further compilation work is required for the 5N Zone to determine if further drilling is required there next, or whether further surveys are required in preparation for potential drilling. According to Wild (2002), to the east, soil sampling, is warranted on some of the recently established grid on the east half of the project area. Also to the east, a coincident max-min conductor and subtle gravity anomaly on lines 3800S to 4700S should be followed up by extending the grid and extending the gravity coverage to adjacent lines. Drilling should be planned on the most promising part of the anomaly (Wild, 2002). Similarly, Walcott concluded that the residual gravity anomalies obtained in relation to EM conductors here be investigated by drilling (Walcott, 2002).

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

Frank Creek Volcanogenic Massive Sulphide Project

Geological mapping should continue in order to help determine the stratigraphic and structural setting of the mineralization. This should be integrated with that being done by Ferri of the B.C. Geological Survey between the Ace and Frank Creek areas. Independent geological consultants of Strathcona Mineral Services (Toronto, Canada) toured the Frank Creek Project and after inspecting core from the drill program of 2002 recommended further work including establishing survey grids, mapping and soil sampling and geophysical surveys similar to those recommended for the Ace Project, followed by trenching and drilling. They concluded that there is still the possibility that massive sulphide mineralization exists on the property and that more work is warranted on the remaining targets.

Barker Minerals has enlisted the services of Dr. Tim Barrett and Dr.Wallace MacLean of Ore Systems Consulting to determine the stratigraphic and alteration relations of the rock sequences hosting the massive sulphide mineralization on the Ace and Frank Creek properties using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography. Ore Systems Consulting has previously worked on numerous volcanic-associated massive sulphide deposits, including those in the Noranda and Matagami camps of Quebec, the Timmins area of Ontario, the Cordillera of British Columbia, Sweden, Wales, Portugal and the Philippines, and have published numerous research papers on these areas, thus this author recommends that they, based on their worldwide experience and advance level of geological expertise in the field of volcanogenic massive sulphide deposits and models, determine the paleotectonic and paleoseafloor setting of the mineralization in each area, as this will have a direct bearing on the VMS model used during exploration. This author also recommends work to determine the chemical identification of the main mineral host rock units at the Ace and Frank Creek Projects in order to trace these favorable units laterally during exploration. The quantitative determination of the degree of alteration is also recommended in order to determine the most intense portions of the hydrothermal systems on each property, as those areas containing a combination of favorable stratigraphy and strong hydrothermal alteration have the greatest potential for economic VMS mineralization. The results should be combined with known geophysical and soil-geochemistry anomalies established both by earlier surveys and potential 2003 surveys in order to select high-priority drilling targets.

The HLEM survey should be continued in areas of the grids not yet covered, especially to trace continuations of known anomalies that are suggestive of the presence of massive sulphide deposits. Additional preliminary gravity survey lines should be run across the main anomalies to more accurately define massive sulphide targets. Where topography and glacial cover are permissive, trenches should be excavated into prospective targets resultant from these studies. VLF geophysical surveys should be used to trace mineralization once it has been located in bedrock.

Geochemical soil surveys should be conducted over the strike extensions of known mineralized trends and also over specific target areas in order to identify base and/or precious metal soil anomalies, which may be indicative of economic massive sulphide mineralization targets in the local bedrock.

The most prospective anomalies as defined by geology, geophysics, geochemistry and trenching should be tested by diamond drilling.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

SCR Project

Geological mapping should continue in order to help determine the stratigraphic and structural setting of the mineralization. This should be integrated with that being done by Ferri of the B.C. Geological Survey between the Ace and Frank Creek areas.

Lithogeochemistry studies should be continued in order to determine the stratigraphic and alteration relations of the rock sequences hosting the known mineralization, using advanced lithogeochemical methods, combined with new core and outcrop sampling and petrography.

The HLEM survey should be continued to the east and southeast of the present grid to determine the extent of the anomaly in those directions. Where permissive, trenching should be conducted on defined target areas.

Soil geochemistry should be conducted over the grid and should be extended to the east to cover the possible extensions of the anomalous zone as defined by geology and coincident magnetic and HLEM anomalies.

Diamond drilling should test the most prospective anomalies as defined by the geology, geophysics, geochemistry and trenching.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

Peacock Showing

A mineral occurrence was found by a BC Geological Survey geologist during his local geological mapping work may be the previously known Peacock mineral showing. The Company should further expose and sample this mineralization. This occurrence has been interpreted by government geologists (Ferri 2001) to be Besshi-type VMS mineralization. All available previous exploration results should be compiled, interpreted and, if then warranted, be followed up with an initial program designed to identify and develop drill targets having economic VMS and/or gold/silver potential.

Blackbear Project Area

Since it appears that interest in gold is increasing as its price has risen substantially during the previous year, the Company should begin to investigate this project area, which hosts the former Providence Mine, from which a previous operator's shipment of selected ore from the No. 2 zone assayed 3343 grams of silver per tonne, 45.7% lead, 0.11% zinc and 4.9 grams of gold per tonne (all the above from BC MinFile 093A 003). One grab sample (# 11-07-98-59; Barker Minerals) of outcropping mineralization contained 52% Pb, 0.03% Zn, 142 oz Ag / ton and 0.081oz Au / ton. (Payne, 1999; BC Assessment Report). All previous exploration results available should be compiled, interpreted and, if then warranted, be followed up with an initial program designed to identify and develop drill targets having economic VMS and/or gold/silver potential.

Cariboo Prospect and Other Areas

Detailed geological mapping should be continued in the Cariboo prospect area in order to help determine the extent of deformation and in order to explore for targets of Zn+Pb replacement deposits along strike of the known zones. Compilation of all relevant data and limited diamond drilling is recommended in order to confirm the previous operator's drilling and in order to further define and investigate the size and economic potential of this deposit, which is open in both directions along strike and at depth. Despite low gold concentrations found there to date, further prospecting in the near vicinity and exploratory investigation of the Foster zone for its gold potential is recommended in light of its position within the Pleasant Valley Thrust and the intensity of sulphides in the zone and their localization at a junction of two large scale structures.

Elsewhere in the eastern half of the property, detailed mapping and follow-up geochemical sampling and geophysical surveys should be performed in areas of significant, multi-element stream-sediment anomalies. Areas of particular interest are the upper reaches of the Sellers Creek and Grain Creek drainage basins.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

Quesnel Platinum Project

Stream and soil sampling programs should be conducted in drainages associated with known placer occurrences of PGE minerals, especially in areas of mafic and ultramafic rocks that may be sources of such minerals. Studies should be done in order to characterize the assemblages of PGEs, which could help determine the environment of their origin. Geological mapping should be conducted in areas of mafic and ultramafic rocks in order to better understand the nature of these rocks and their possible association with PGE minerals. Prospecting should continue in drainages that contain known placer Pt deposits or anomalous concentrations of PGEs.

The zone of large copper concentrations in the Mag claims should be studied in more detail geologically, geochemically and geophysically. Other zones of coincident geochemical and geophysical anomalies should be examined once data from previous reports are compiled and interpreted. Some of these areas will require new grids for geophysical and geochemical surveys.

The second phase of the program will be contingent on obtaining sufficient positive results from the first phase of the proposed program.

Exploration Budget

September 1, 2002 - November 30, 2003

Ace

Phase 1

Geological mapping and supervision	$ 25,000
Grids, line-cutting, surveying	$ 25,000
Trenching	$ 50,000
Geophysical surveys	$ 50,000
Geochemical surveying, sampling and assaying	$ 30,000
Petrographic and mineralogical studies	$ 10,000
Diamond Drilling (4,000m @ $100/m**)	$ 400,000
General Field	$ 40,000
Total	**$ 630,000**

*** Phase 2**

Geological mapping and supervision	$ 50,000
Grids, line-cutting, surveying	$ 50,000
Trenching	$ 50,000
Geophysical surveys	$ 100,000
Geochemical surveying, sampling and assaying	$ 100,000
Petrographic and mineralogical studies	$ 25,000
Diamond Drilling (10,000m @ $100/m**)	$ 1,000,000
General Field	$ 100,000
Total	**$ 1,475,000**

Total Phase 1	**$ 630,000**
Total Phase 2	**$ 1,475,000**

* Phase 2 will be contingent on obtaining sufficient positive results from Phase 1.

** $100/m all-in, including mobe/de-mobe, drilling, core logging, splitting, sampling and assaying.

Frank Creek & SCR

Phase 1

Geological mapping and supervision	$ 8,000
Grids, line-cutting, surveying	$ 8,000
Trenching	$ 10,000
Prospecting	$ 5,000
Geophysical surveys	$ 20,000
Geochemical surveying, sampling and assaying	$ 3,000
Petrographic and mineralogical studies	$ 1,000
Diamond Drilling (6,000m @ $100/m**)	$ 600,000
General Field	
Exploration Manager	$ 30,000
Field Manager	$ 17,000
Vehicle rental	$ 15,000
Field management and camp staff costs, groceries	$ 30,000
Administration, report writing, map drafting	$ 8,000
Fuel	$ 5,000
Total	**$ 760,000**

Frank Creek & SCR

***Phase 2**

Geological mapping and supervision	$ 50,000
Grids, line-cutting, surveying	$ 40,000
Trenching	$ 50,000
Geophysical surveys	$ 100,000
Geochemical surveying, sampling and assaying	$ 50,000
Petrographic and mineralogical studies	$ 25,000
Diamond Drilling (10,000m @ $100/m**)	$ 1,000,000
General Field	$ 100,000
Total	**$ 1,415,000**

Total Phase 1	**$ 760,000**
Total Phase 2	**$ 1,415,000**

* Phase 2 will be contingent on obtaining sufficient positive results from Phase 1.
** $100/m all-in, including mobe/de-mobe, drilling, core logging, splitting, sampling and assaying.

Cariboo and Foster Zone

*** Phase 1**

Geological mapping and supervision	$ 25,000
Grids, line-cutting, surveying	$ 25,000
Trenching	$ 10,000
Geophysical surveys	$ 50,000
Geochemical surveying, sampling and assaying	$ 25,000
Petrographic and mineralogical studies	$ 5,000
Diamond Drilling (1,500m @ $100/m**)	$ 150,000
General Field	$ 50,000
Total	**$ 340,000**

*** Phase 2**

Geological mapping and supervision	$ 50,000
Grids, line-cutting, surveying	$ 50,000
Trenching	$ 25,000
Geophysical surveys	$ 100,000
Geochemical surveying, sampling and assaying	$ 25,000
Petrographic and mineralogical studies	$ 10,000
Diamond Drilling (5,000m @ $100/m**)	$ 500,000
General Field	$ 100,000
Total	**$ 860,000**

Total of Phase 1	**$ 340,000**
Total of Phase 2	**$ 860,000**

* Phase 2 will be contingent on obtaining sufficient positive results from Phase 1.
** $100/m all-in, including mobe/de-mobe, drilling, core logging, splitting, sampling and assaying.

Quesnel Platinum/Peripheral

Phase 1

Geological mapping and supervision	$	9,000
Prospecting	$	9,000
Grids, line-cutting, surveying	$	10,000
Geophysical surveys	$	5,000
Geochemical surveying, sampling and assaying	$	2,500
Petrographic and mineralogical studies	$	1,000
General Field		
Exploration Manager	$	4,400
Field Manager	$	2,600
Vehicle rental	$	2,000
Field management and camp staff costs, groceries	$	3,000
Administration, report writing, map drafting	$	1,000
Fuel	$	500
Total	**$**	**50,000**

*** Phase 2**

Geological mapping and supervision	$	15,000
Grids, line-cutting, surveying	$	15,000
Trenching	$	20,000
Geophysical surveys	$	40,000
Geochemical surveying, sampling and assaying	$	25,000
Petrographic and mineralogical studies	$	5,000
Diamond Drilling (900m @ $100/m**)	$	90,000
General Field	$	40,000
Total	**$**	**250,000**

Total Phase 1	**$**	**50,000**
Total Phase 2	**$**	**250,000**

* Phase 2 will be contingent on obtaining sufficient positive results from Phase 1.

Total of all Phase 1	**$**	**1,780,000**
Total of all Phase 2	**$**	**4,000,000**
Total of Phase 1,and Phase 2	**$**	**5,780,000**

Signed on this 21 day of October, 2002

Bruce J. Perry, Ph.D., P. Geo

Effective Date: October 21, 2002.

Certificate of Author

I, Bruce J. Perry, of 2301 Skeena Drive, Kamloops, B. C., V2E 1Y2, do hereby certify that:

1. I have supervised and participated in the preparation of this Report on Exploration of the Barker Minerals Ltd. Property.

2. I am a "qualified person" as defined in National Instrument 43-101: Standards of Disclosure for Mineral Projects ("NI 43-101") and my qualifications include the following:

(a) I graduated from the University of Toronto, with distinction in 1984 with a B. Sc. degree in Geology.

(b) I graduated from the University of Toronto in 1990 with a M. Sc. degree in Geology.

(c) I graduated from the University of Toronto in 1994 with a Ph. D. degree in Geology.

(d) I am a Professional Geoscientist (P. Geo.) registered in the Association of Professional Engineers and Geoscientists of British Columbia, member# 23613, and have been a member in good standing since 1998.

(e) I am a Fellow of the Geological Association of Canada, Fellow No. 6569, and have been a member in good standing since 1997.

(f) From 1987 to the present, I have been actively engaged as a geologist in mineral exploration, mainly in Ontario and British Columbia, Canada.

(g) BC Certified Supervisor of Underground Mines (1995-2002) and BC Certified Supervisor of Open Pit Mines (1995-2002).

3. I have worked on the properties of Barker Minerals Ltd. as a consulting geologist intermittently since 2000 and my most recent visit to the properties of Barker Minerals Ltd. was for two days in October 2002.

4. I am not aware of any material fact or material change with respect to the subject matter of the Report, which is not reflected in the Report, the omission to disclose which makes the Qualification Report misleading.

5. I am independent of Barker Minerals Ltd. based on the tests set out in Section 1.5 of NI 43-101.

6. I have read and NI 43-101 and Form 43-101 F1 and the Report has been prepared in compliance with NI 43-101 and Form 43-101 F1.

Dated October 21, 2002.

Bruce J. Perry, Ph. D., P. Geo., F. G. A. C.

Item 23: References

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Item 26: Figures and Tables

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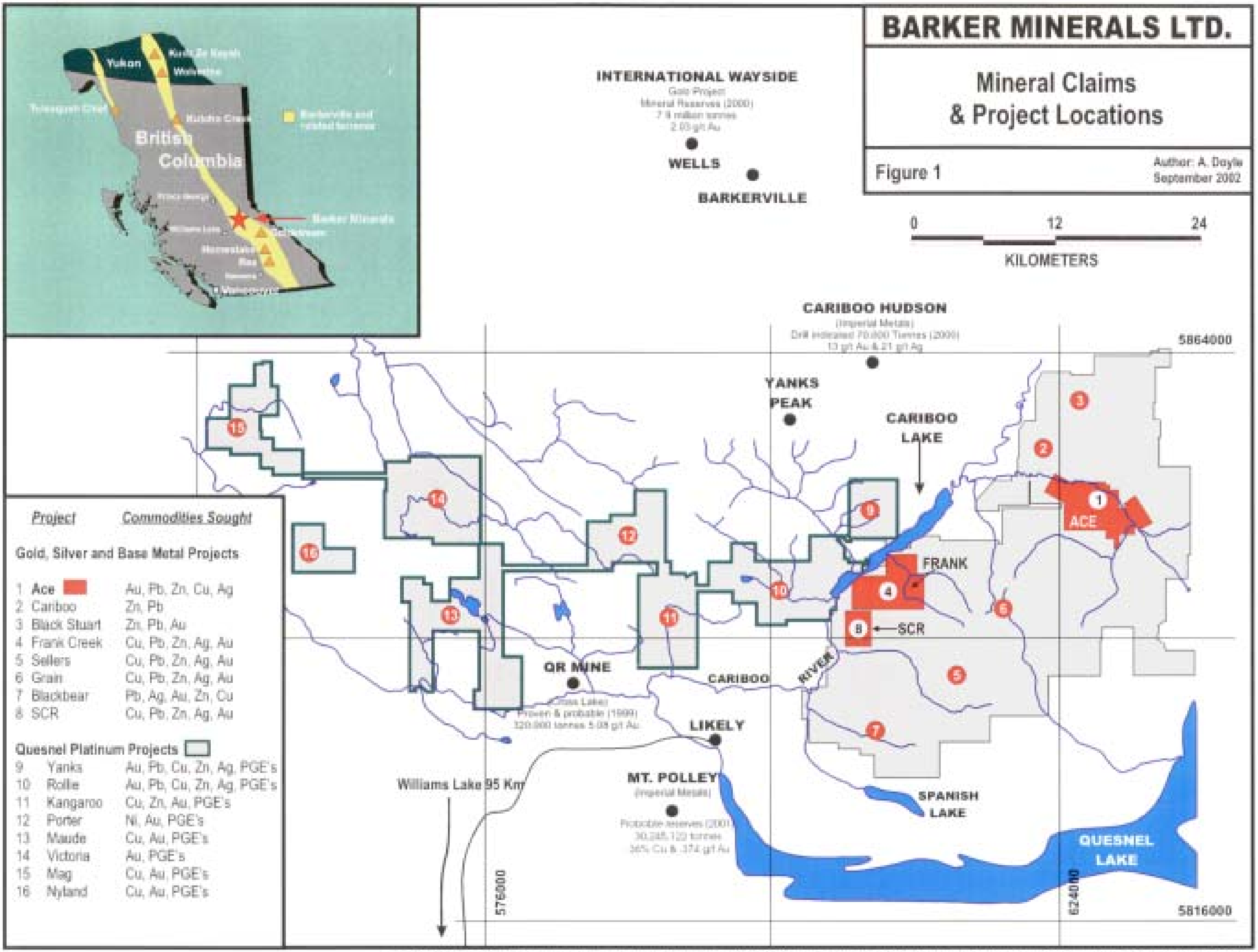

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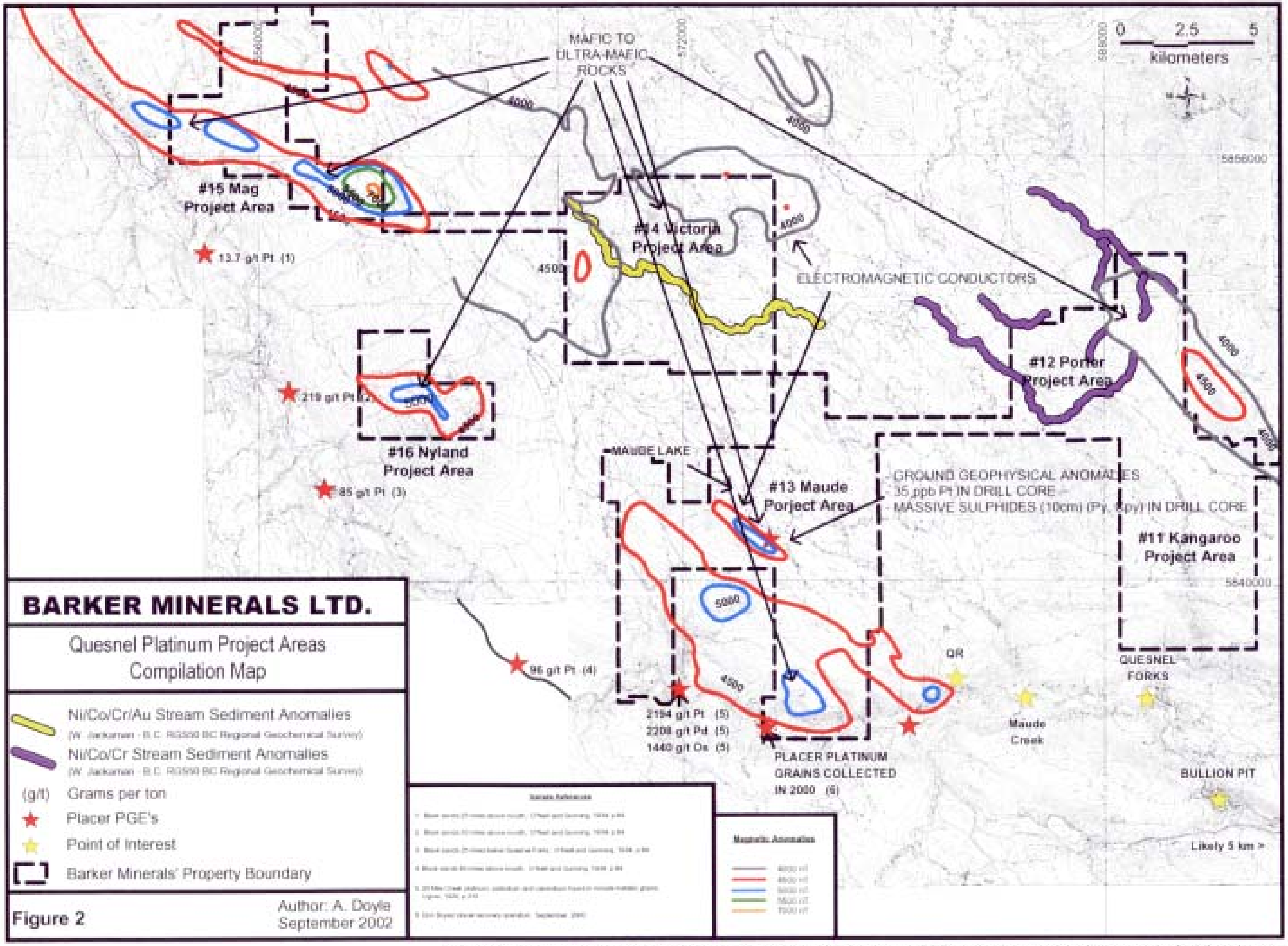

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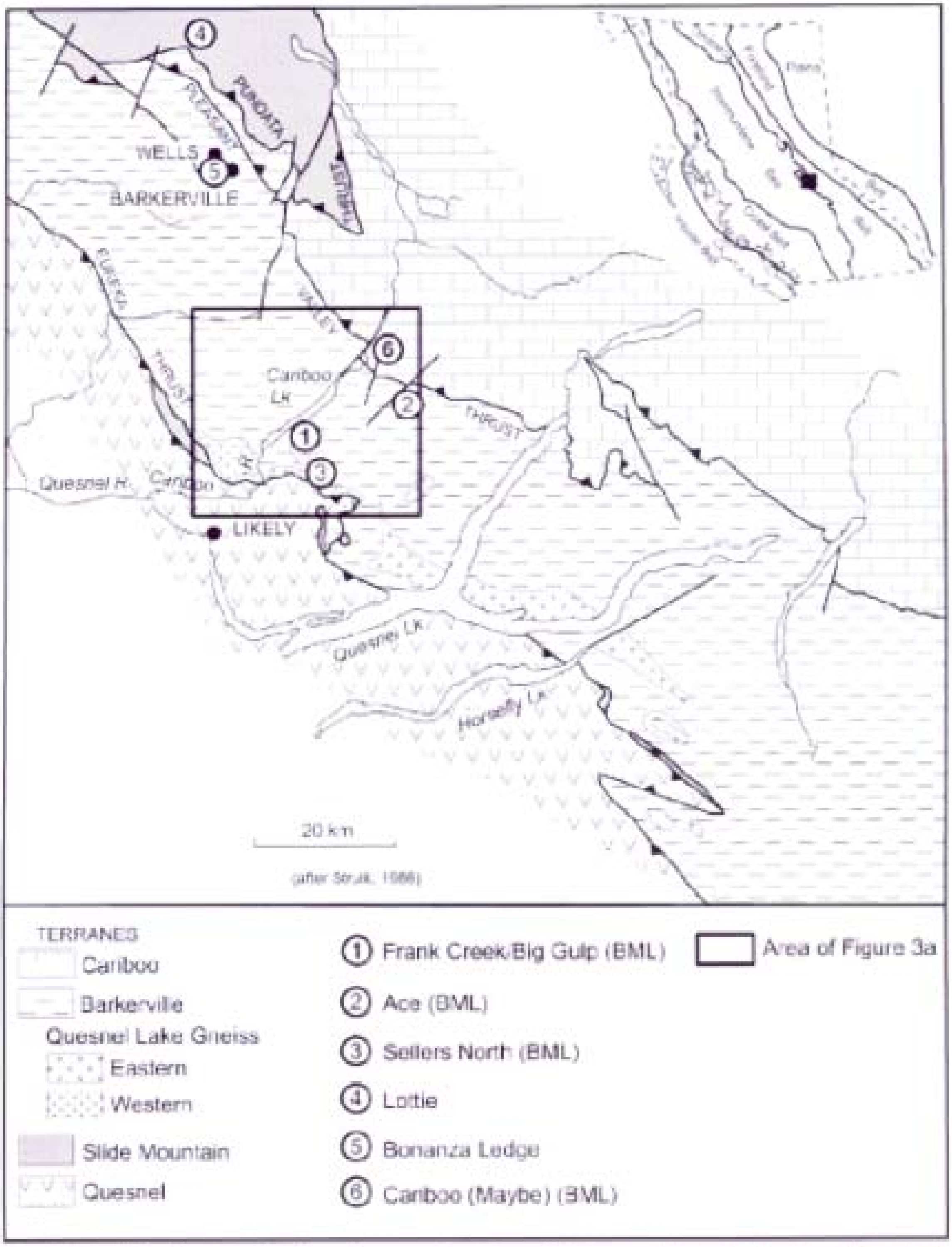

Figure 3. Generalized geological setting of the study area, together with some of the more significant mineral showings in the Barkerville and Cariboo terranes.

BARKER MINERALS LTD.	
Regional Geology (after Ferri, 2002)	
Figure 3	Author: A. Doyle September 2002

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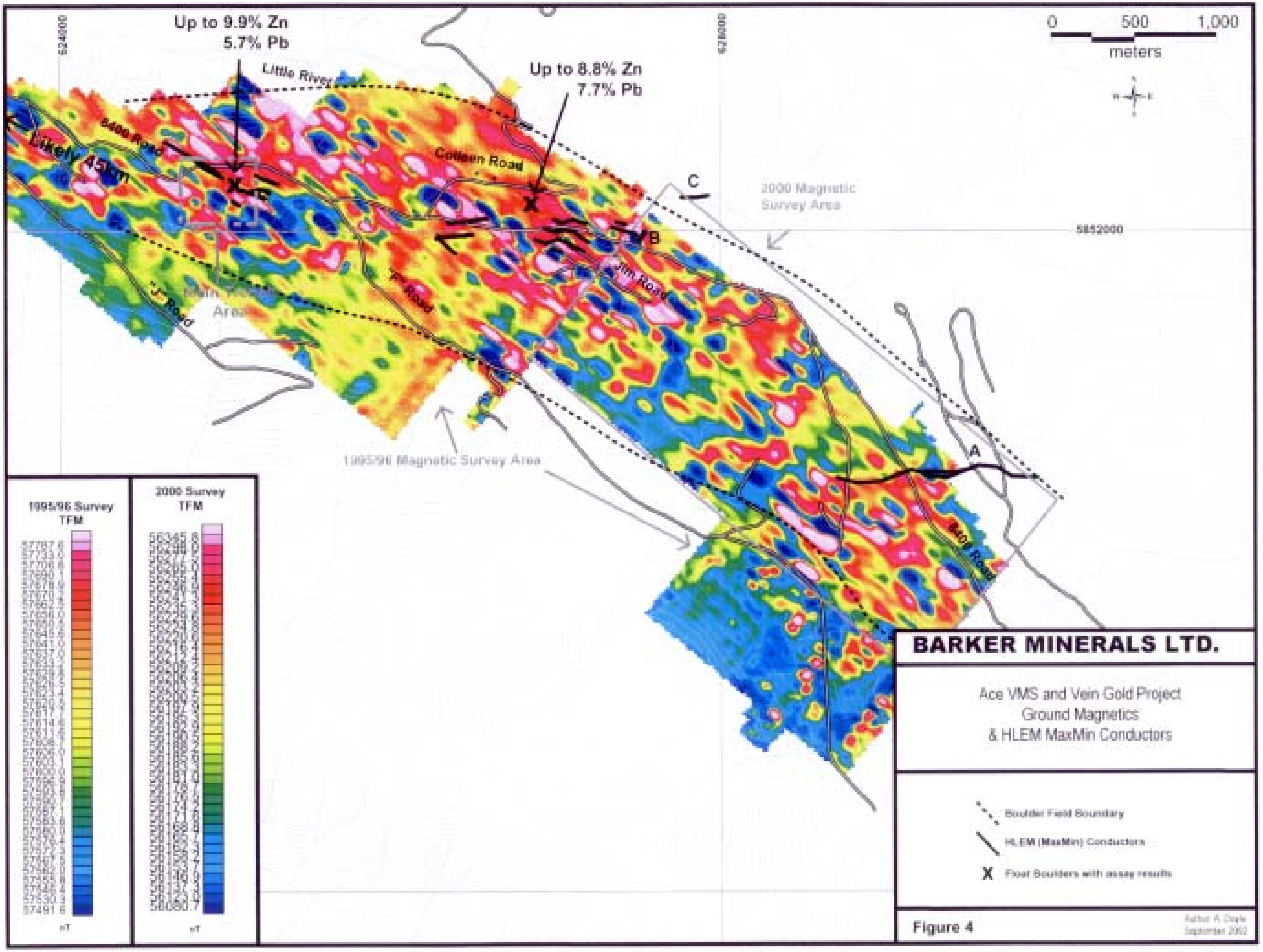

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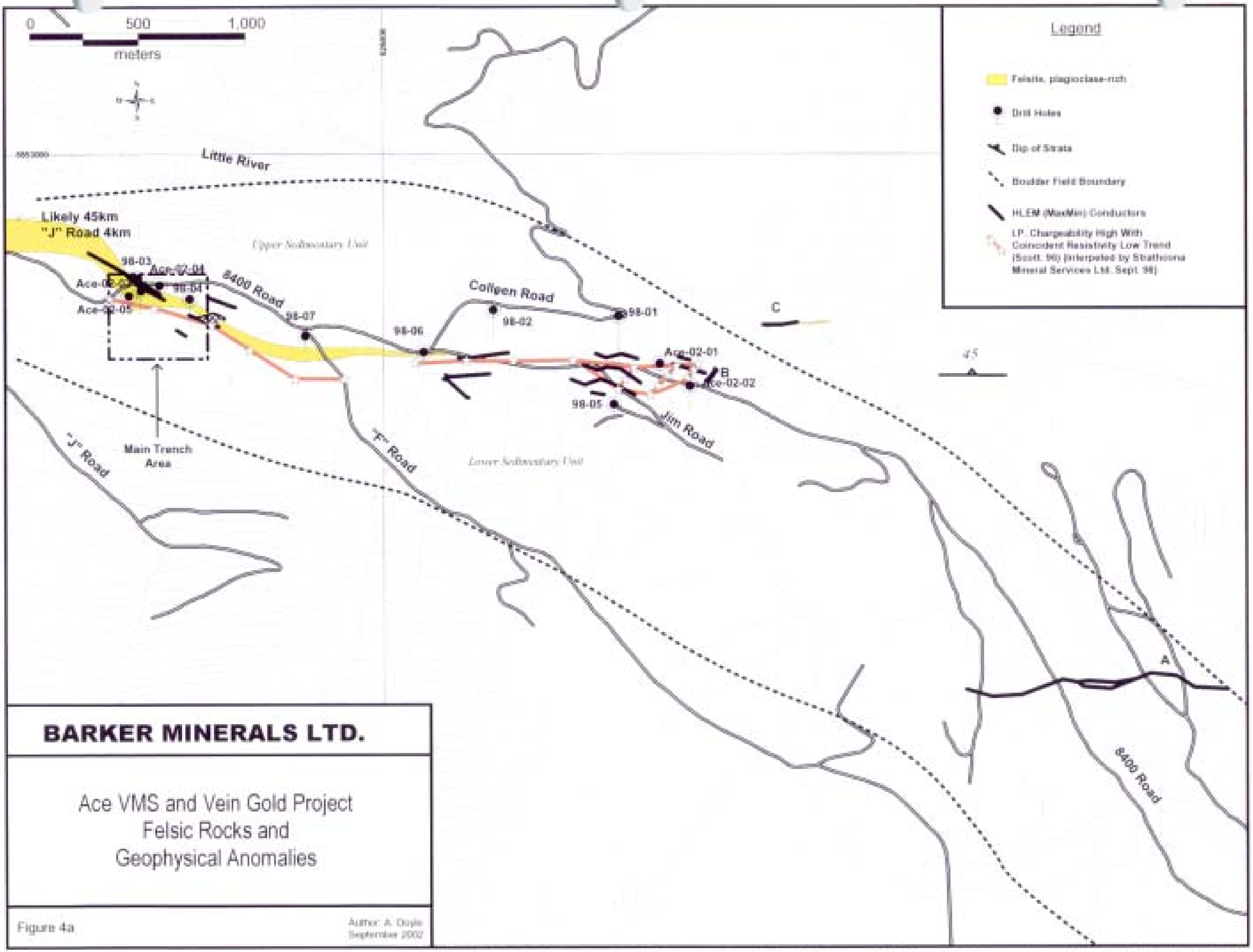

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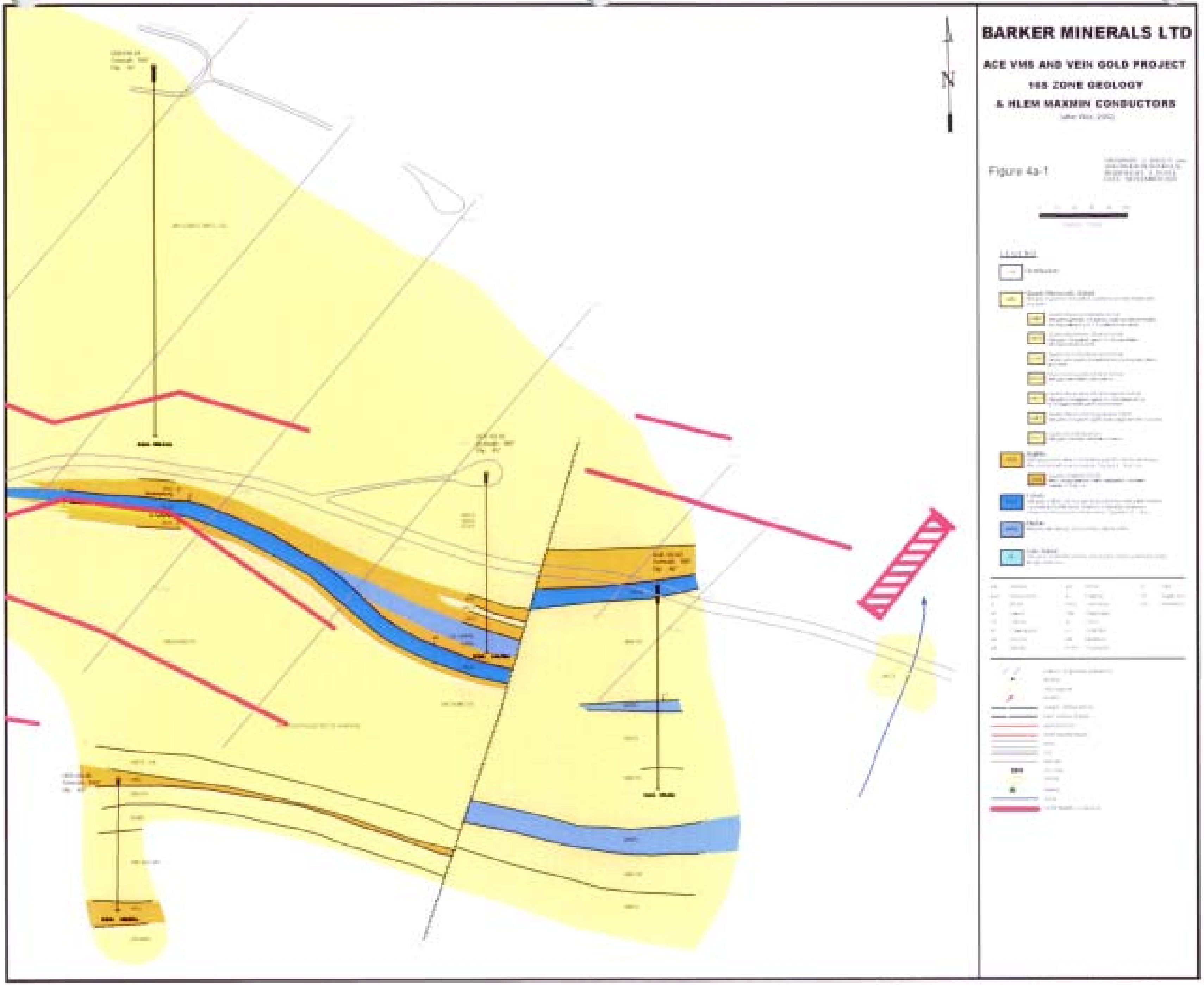

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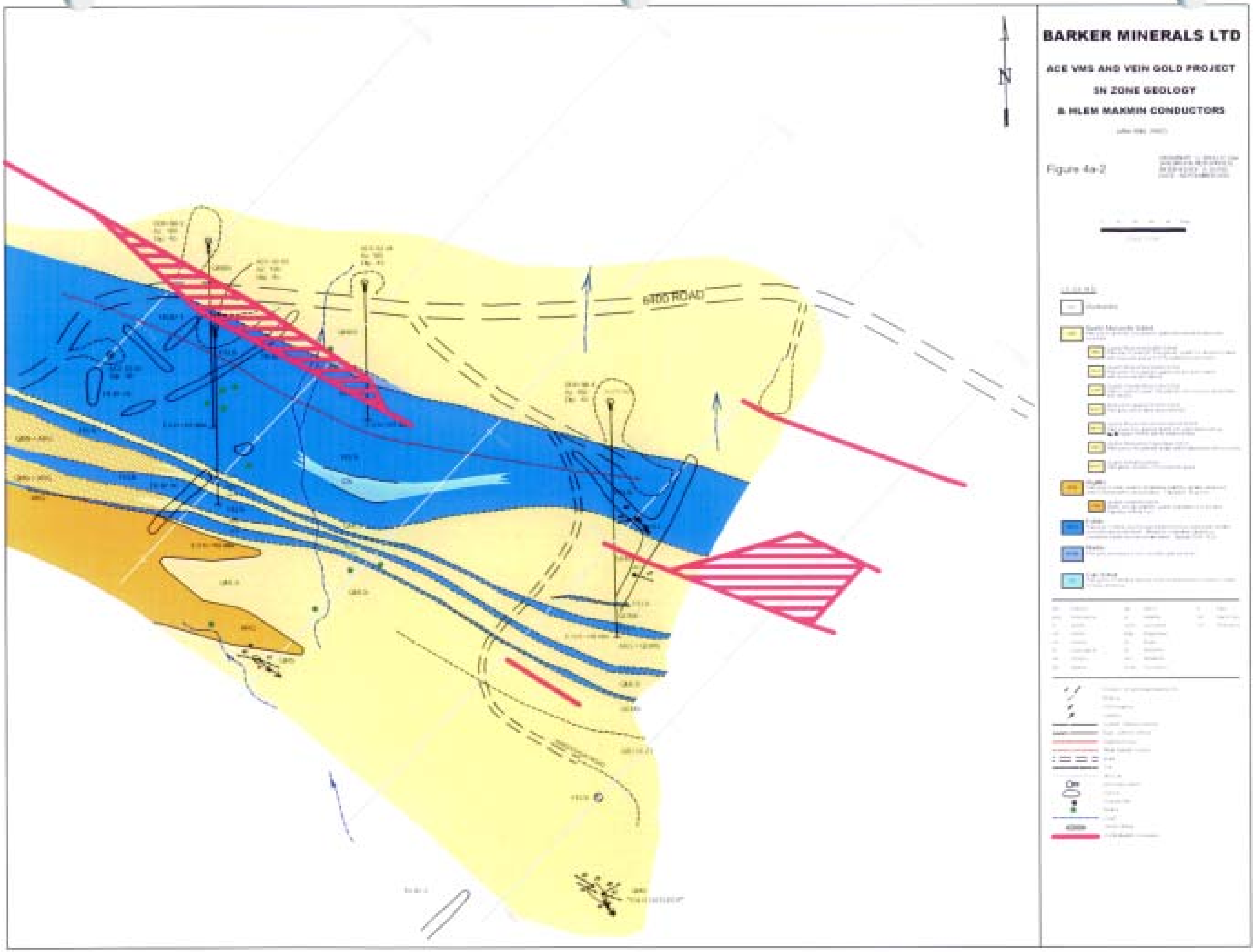

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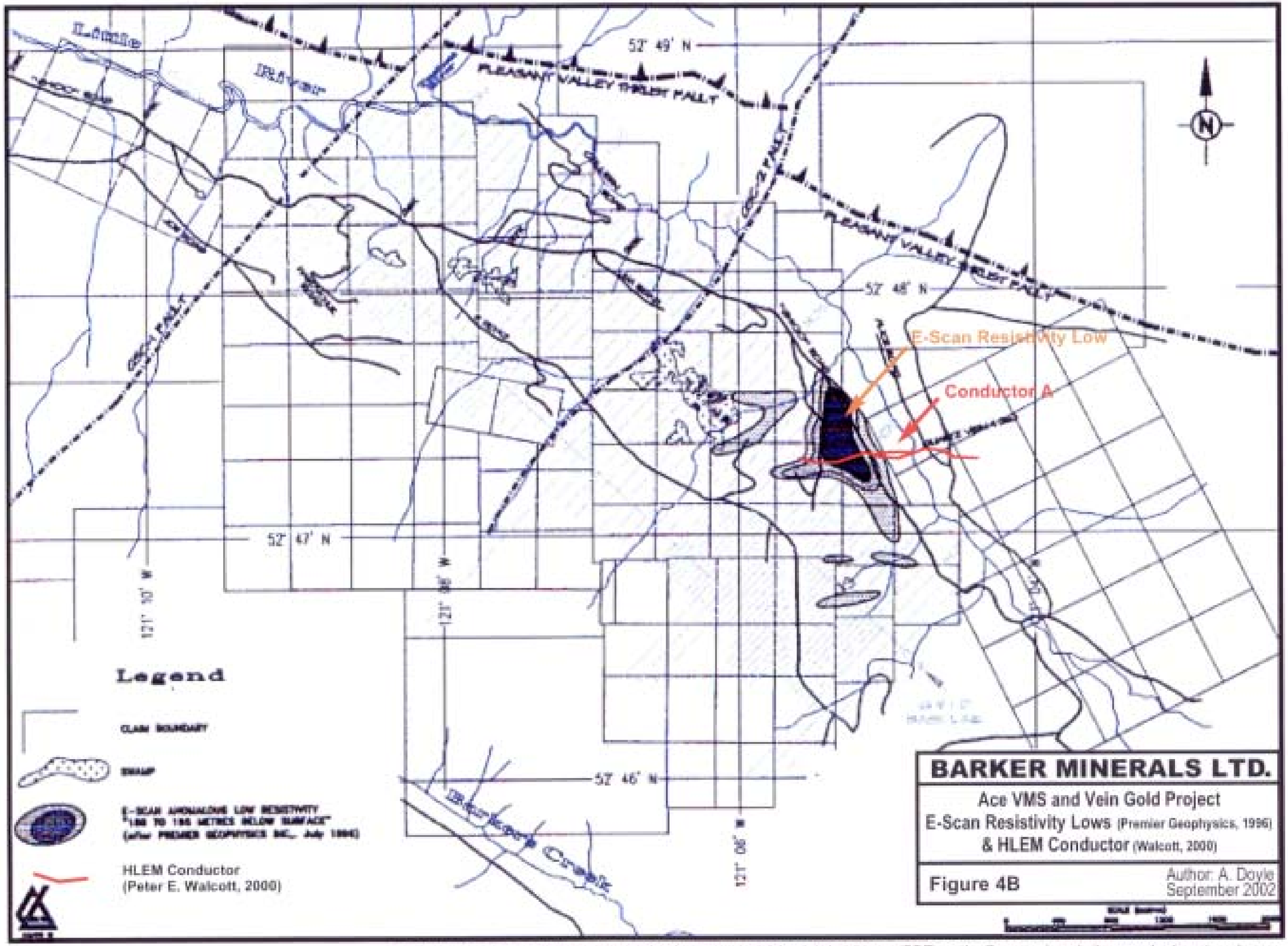

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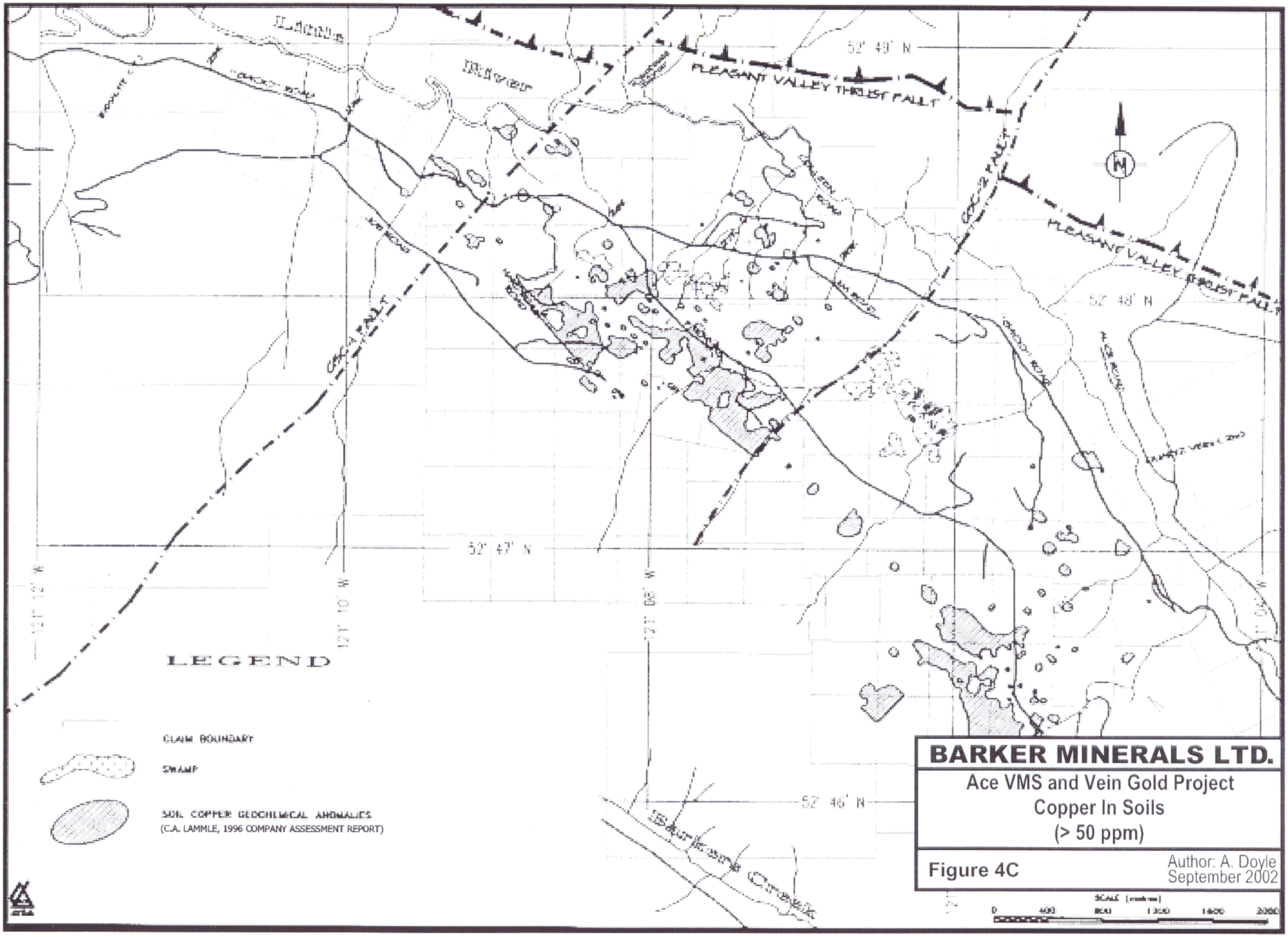

BARKER MINERALS LTD.

Ace VMS and Vein Gold Project
Copper In Soils
(> 50 ppm)

Figure 4C

Author: A. Doyle
September 2002

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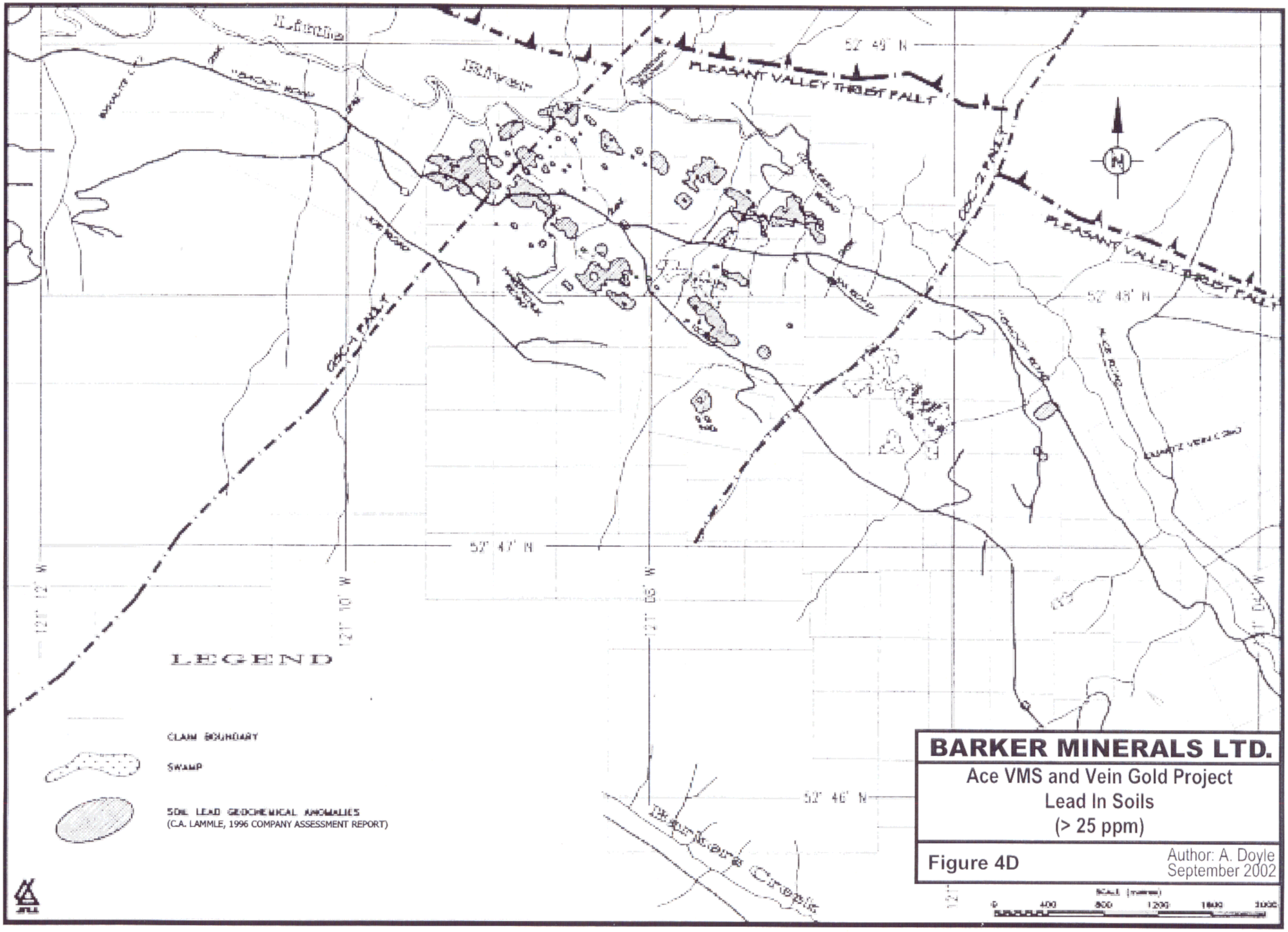

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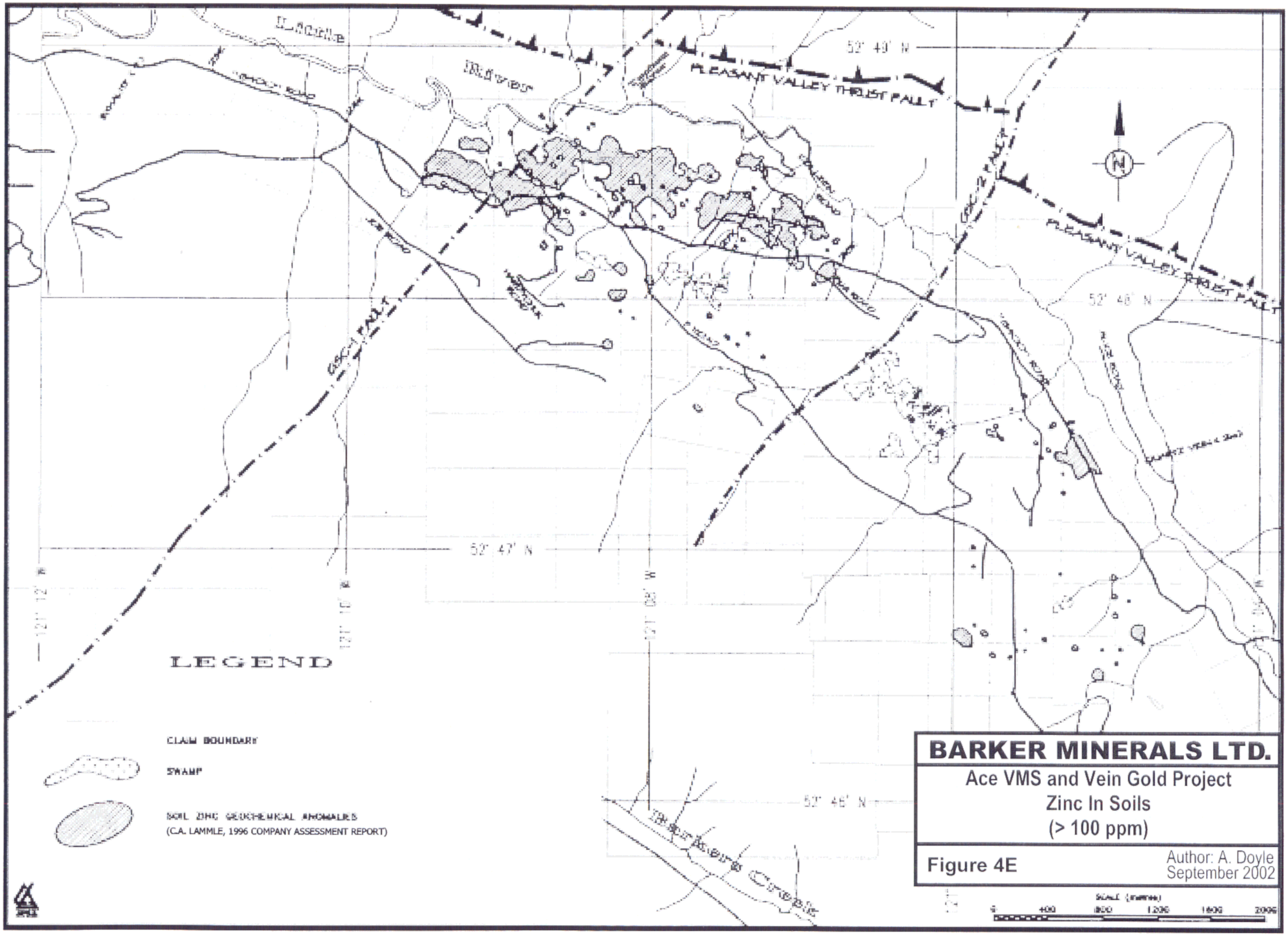

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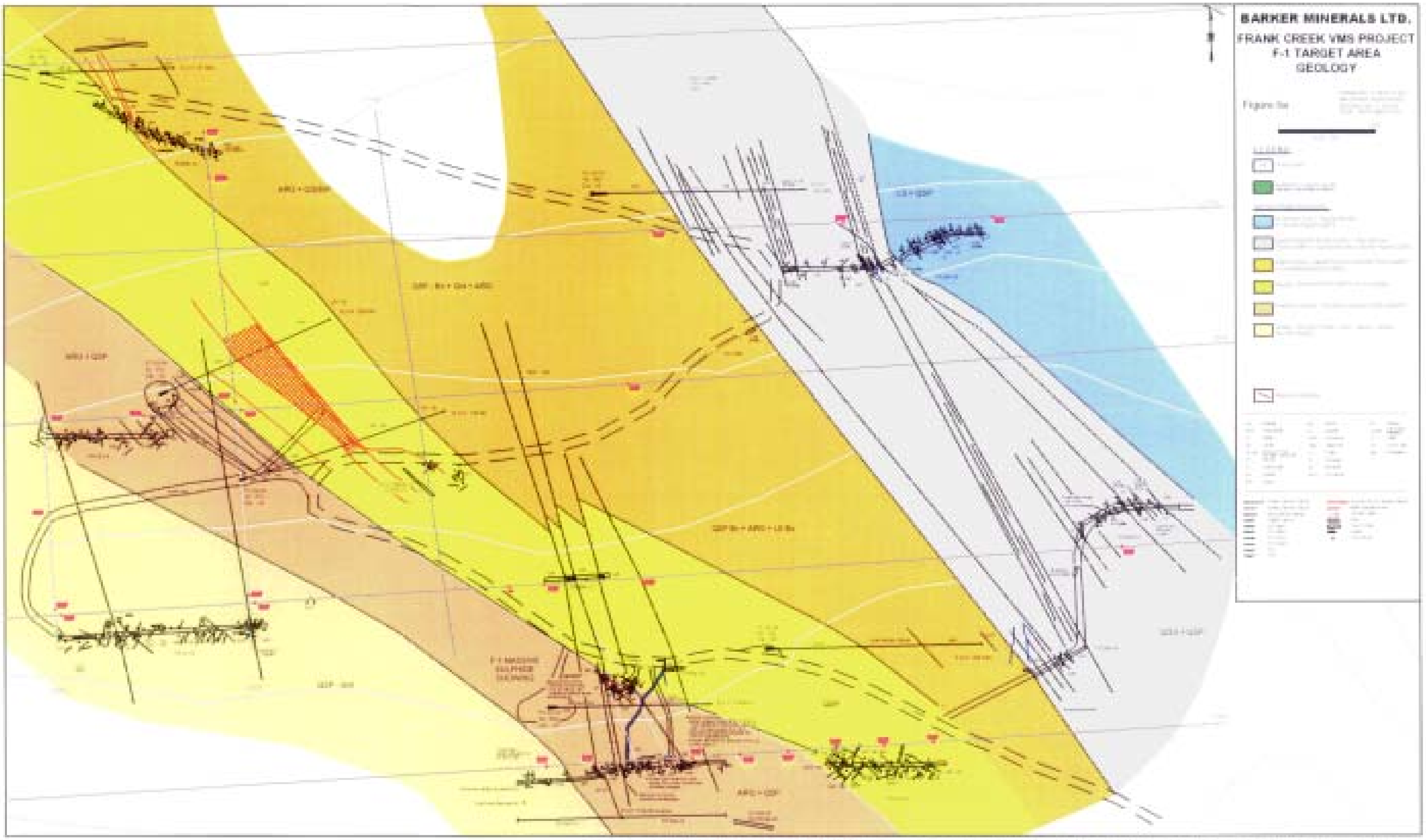

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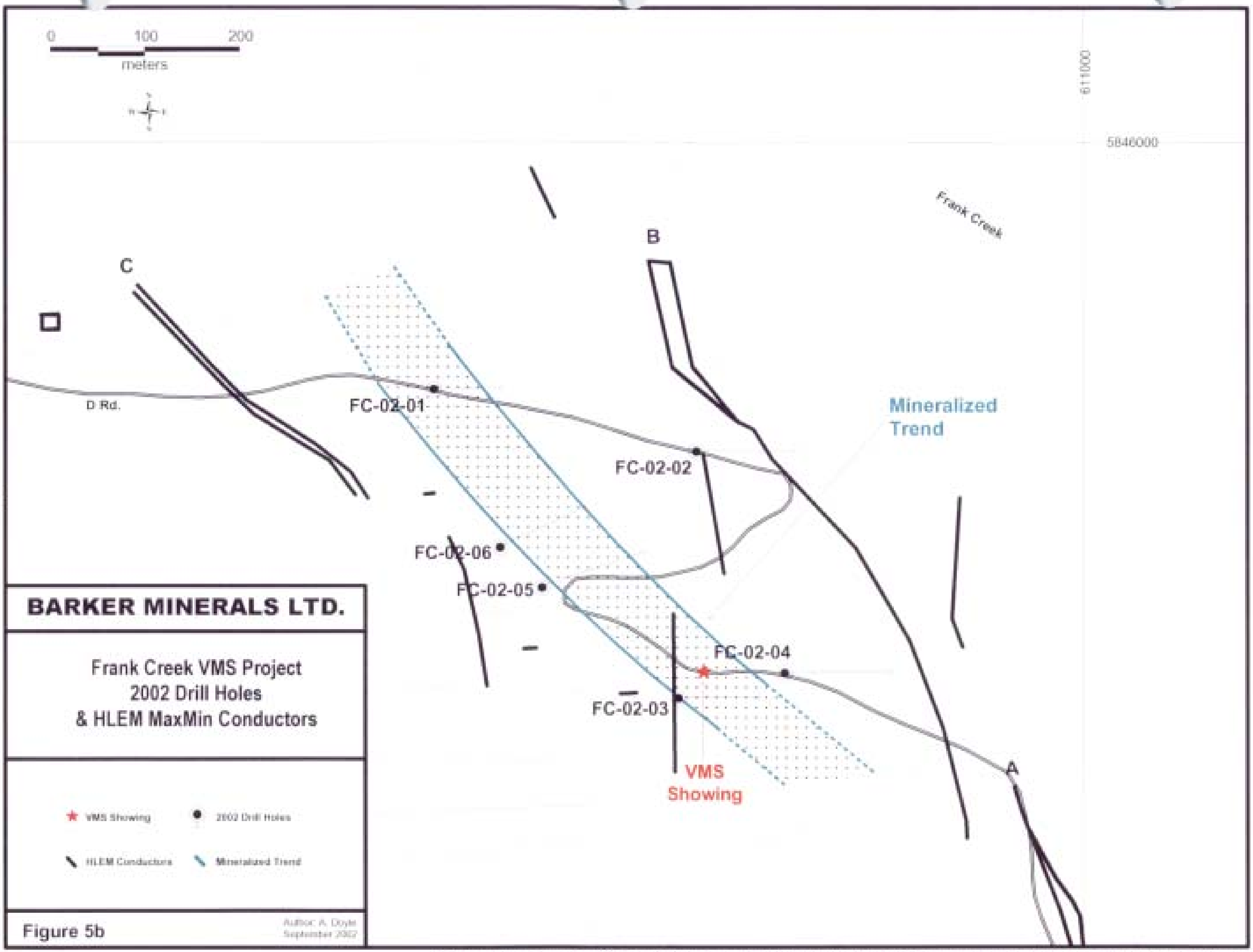

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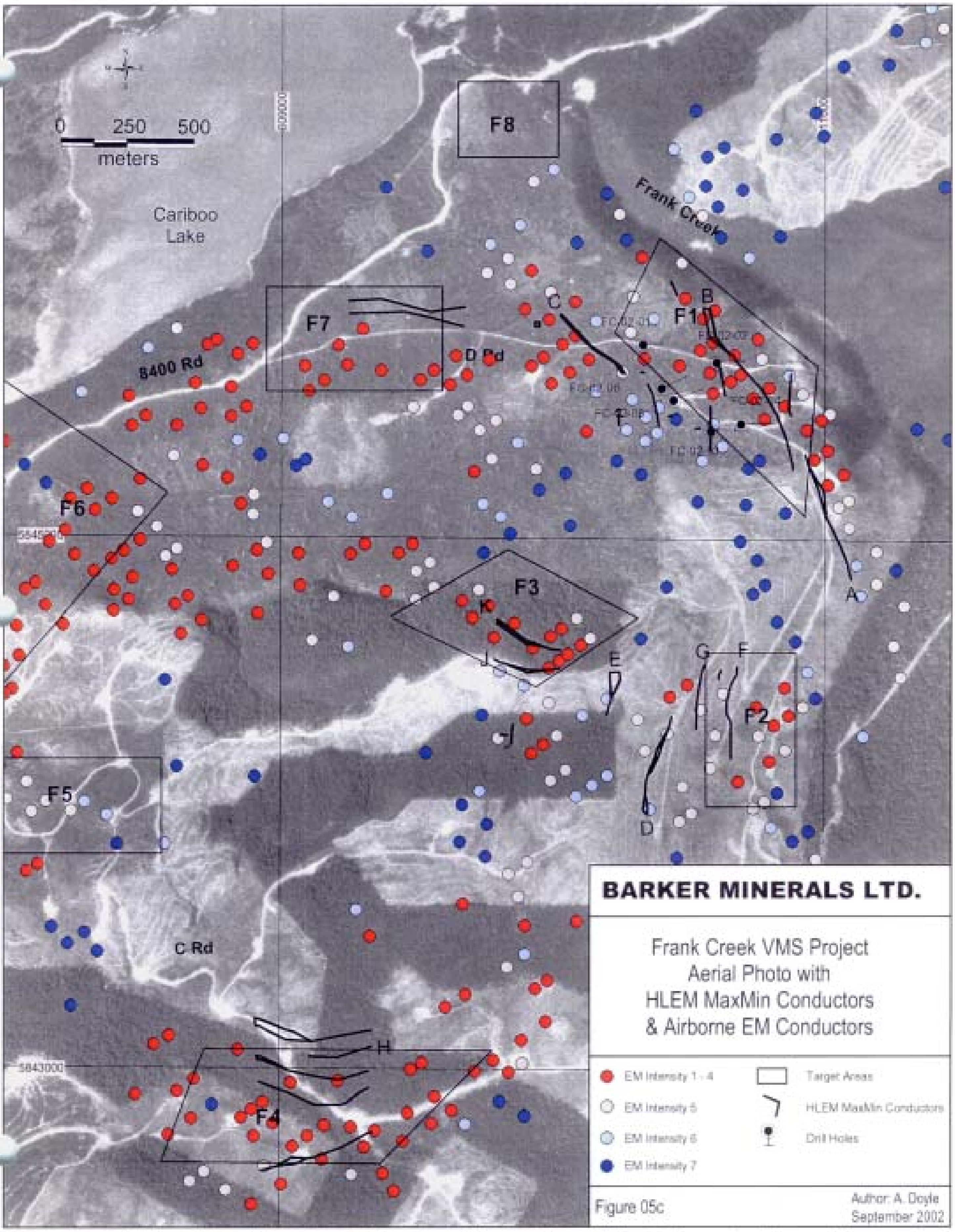

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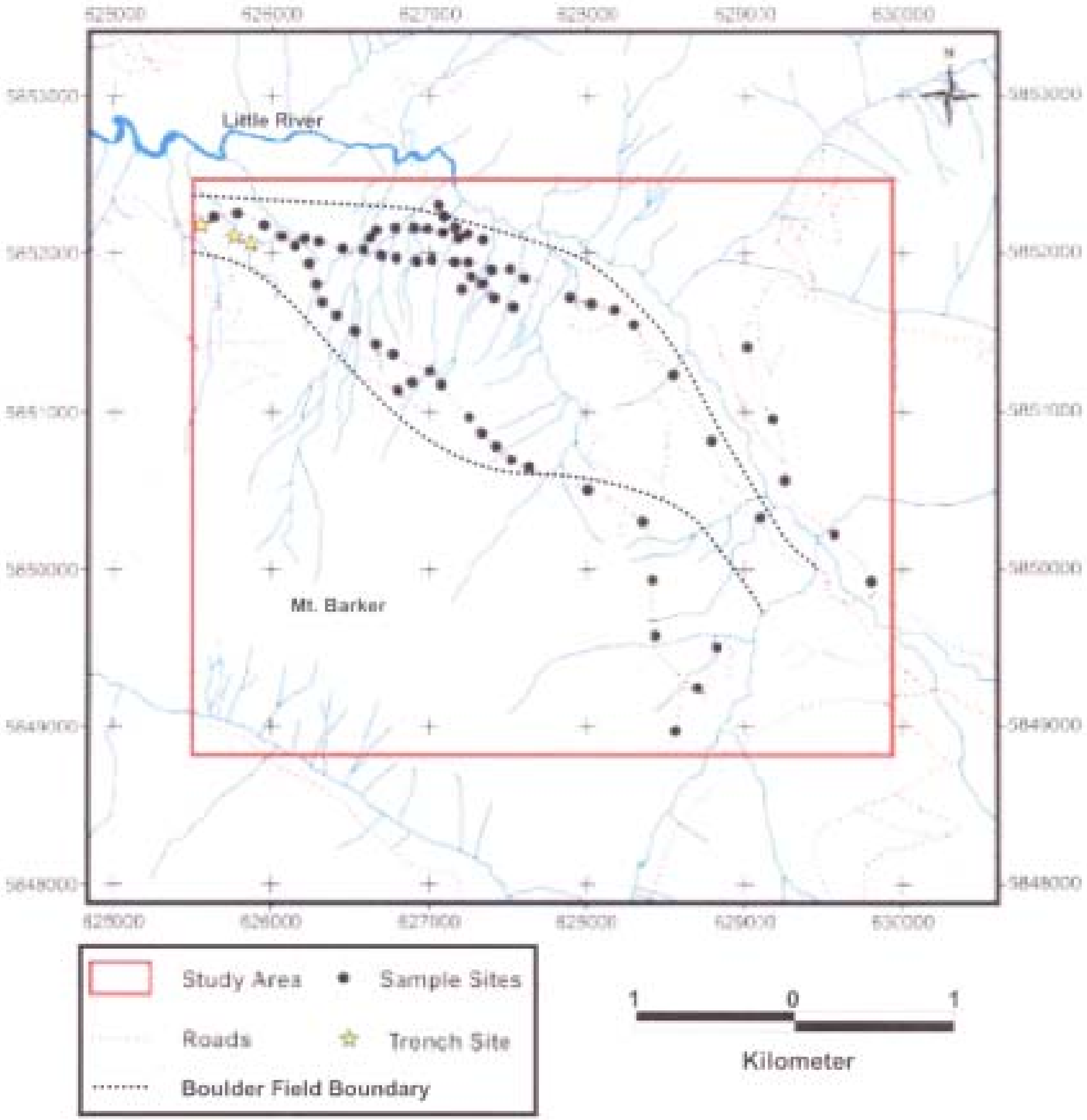

Till geochemistry sample distribution within the Ace Project area [Samples were collected from back-hoe trenches located along roads.]

BARKER MINERALS LTD.

Ace VMS and Vein Gold Project
Glacial Till Study
Sample Locations
(after Bobrowski, BC Geological Survey, 2001)

Figure 6a

Author: A. Doyle
September 2002

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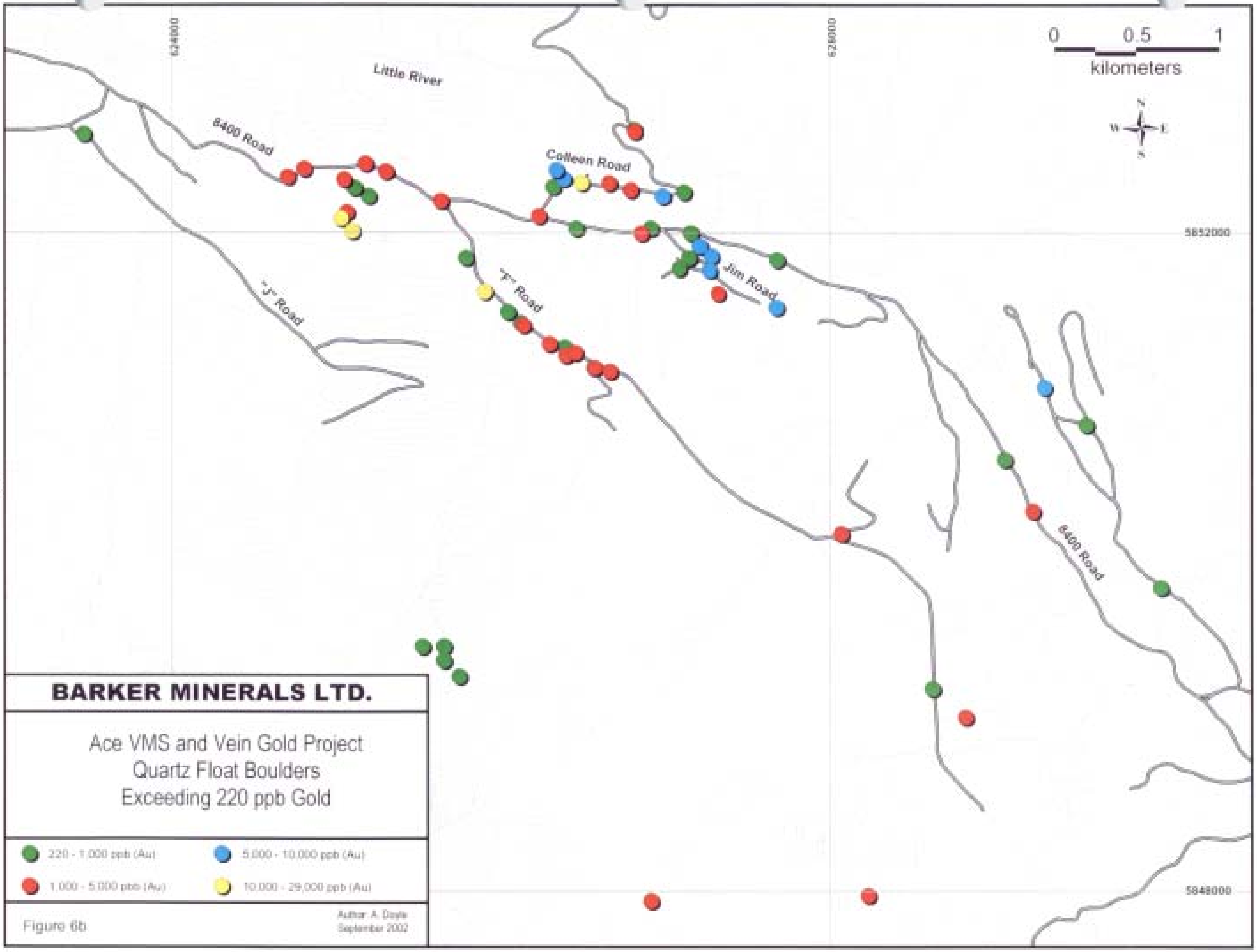

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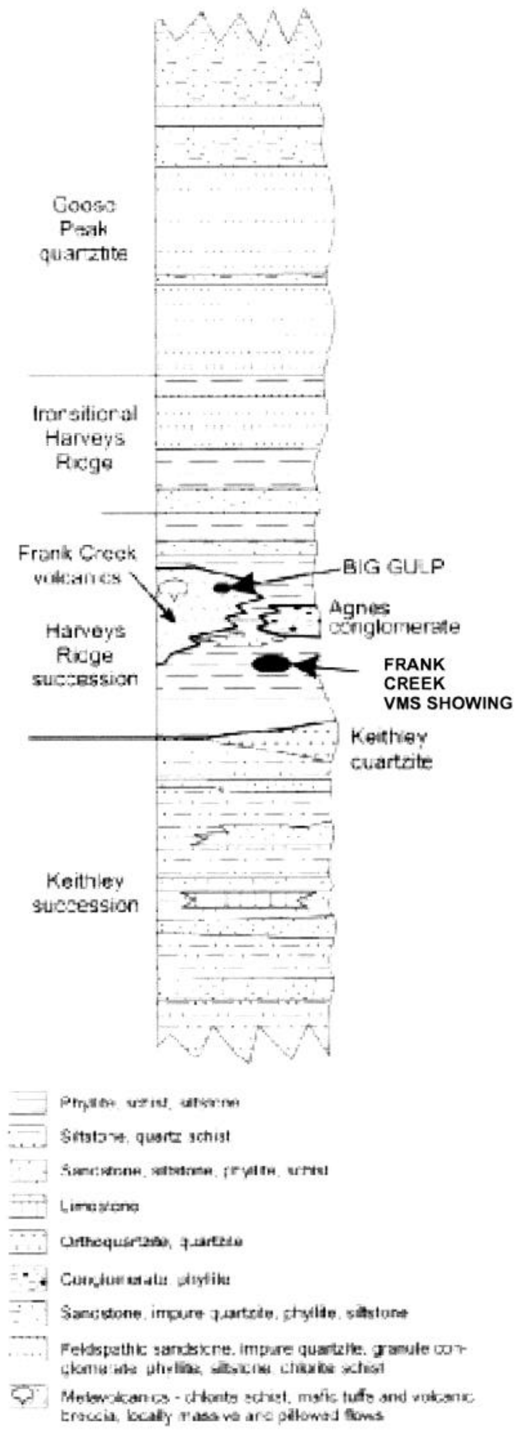

Generalized stratigraphic column of main rock units found within the map area.

BARKER MINERALS LTD.	
Generalized Stratigraphic Column (after Ferri, 2001)	
Figure 7a	Author: A. Doyle September 2002

* Enlargeable detailed figures are available for review in Adobe Acrobat format (PDF) on the Company's website: www.barkerminerals.com

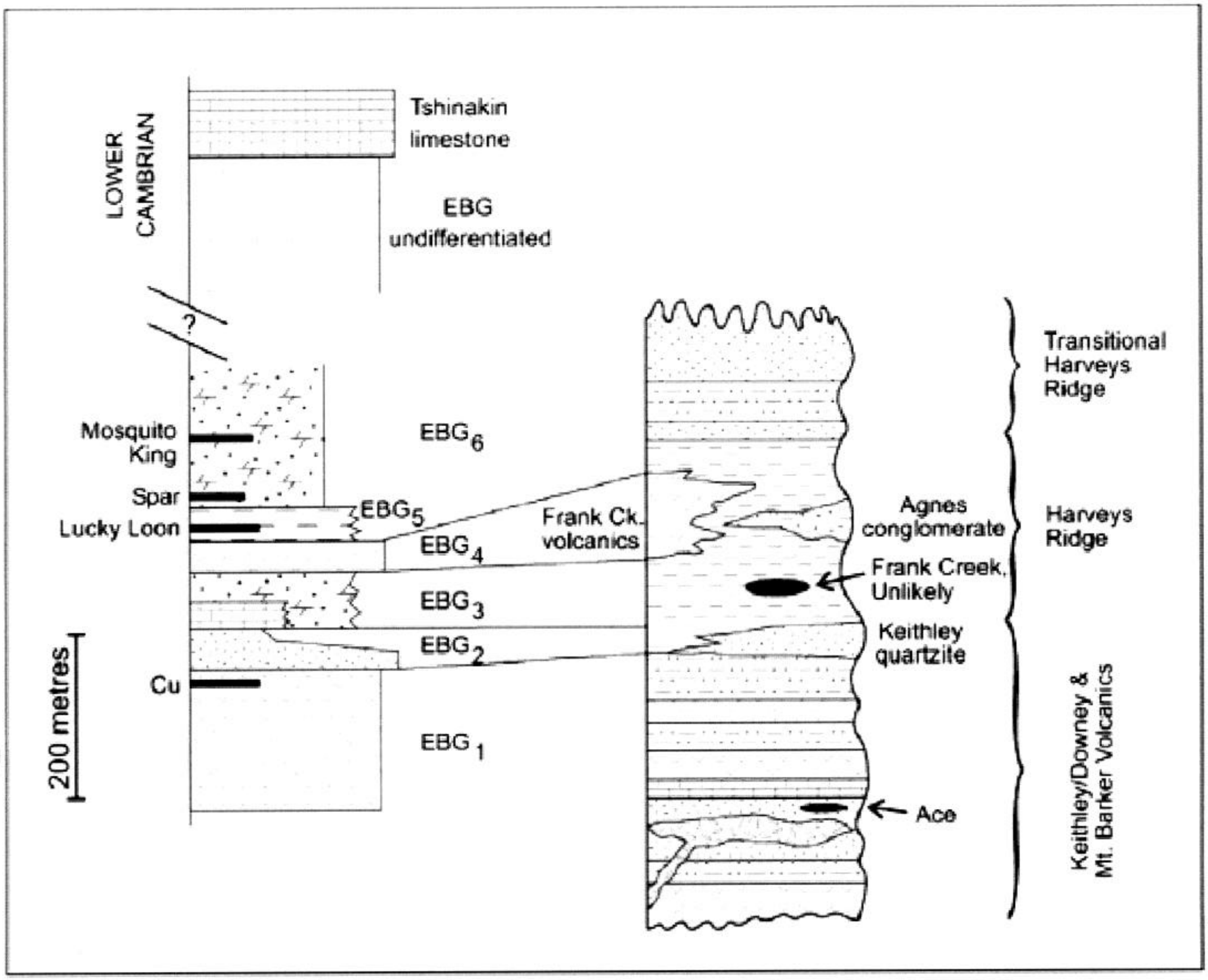

Figure 7b. Correlation of EBG stratigraphy associated with Lucky Coon and Mosquito King mineralization on the Adams Plateau (Höy, 1999) and proposed stratigraphy and associated mineralization within the Cariboo Lake map area.

BARKER MINERALS LTD.	
Barkerville and Kootney Terrane Correlations (after Ferri, 2002-1)	
Figure 7b	Author: A. Doyle September 2002

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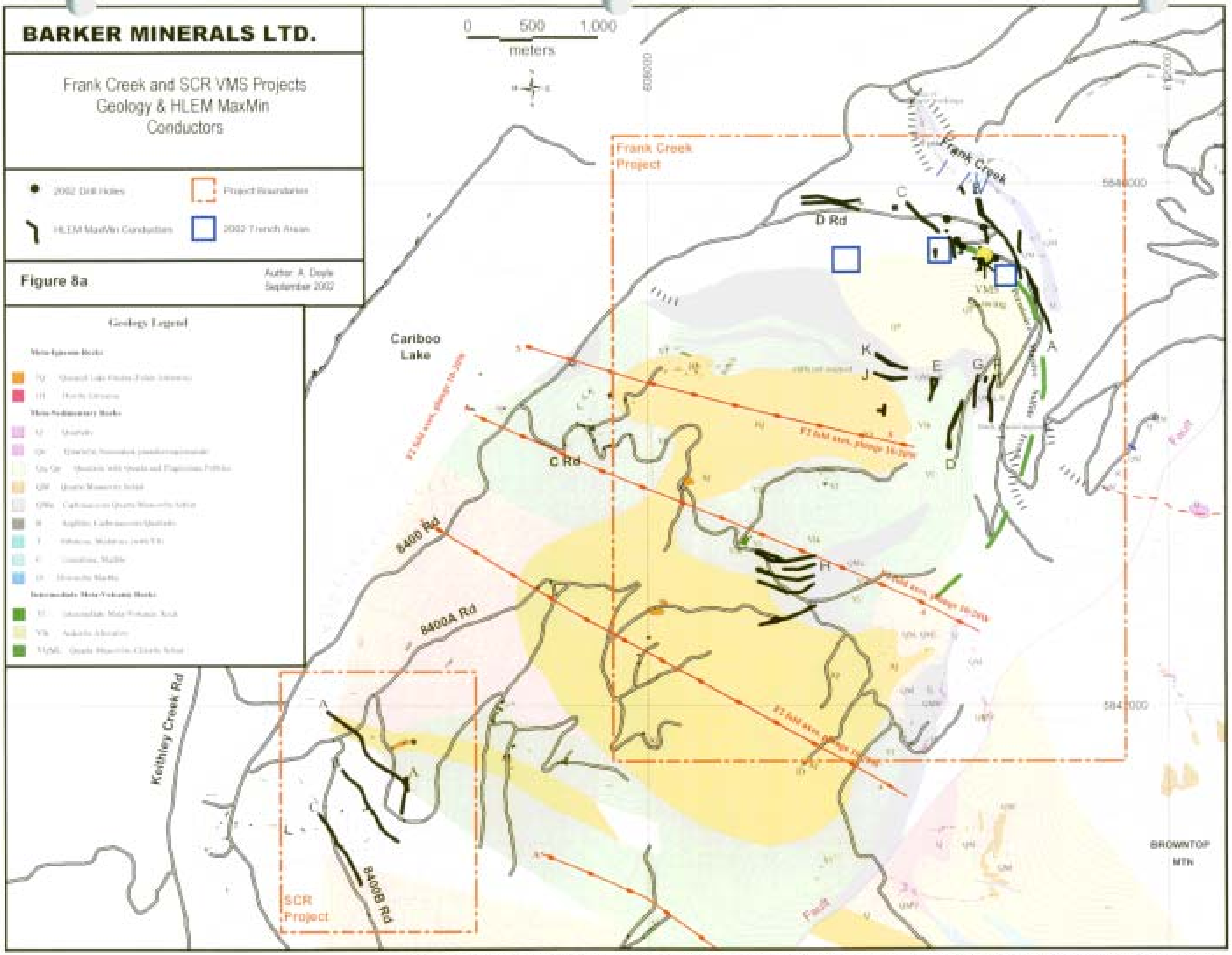

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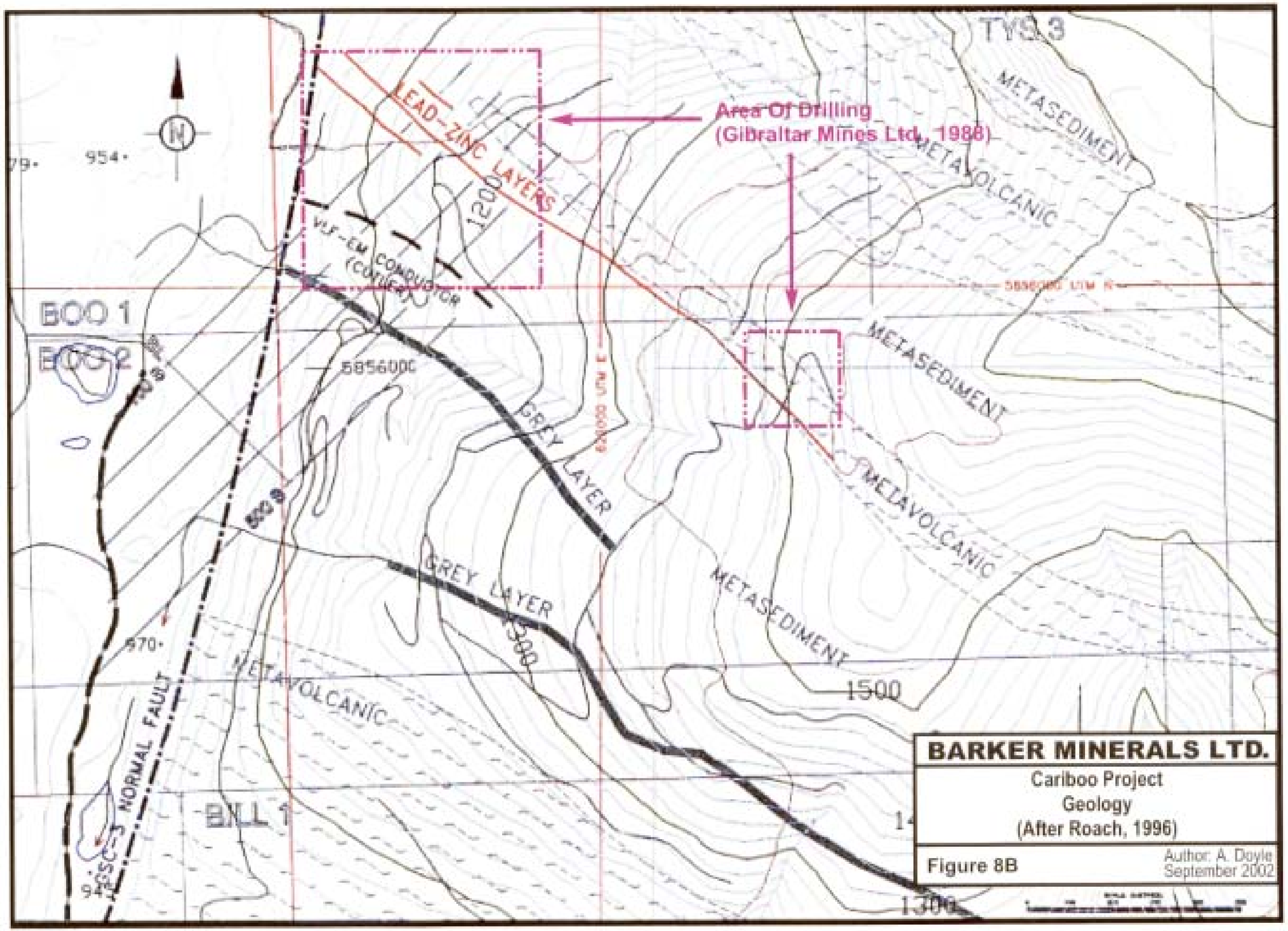

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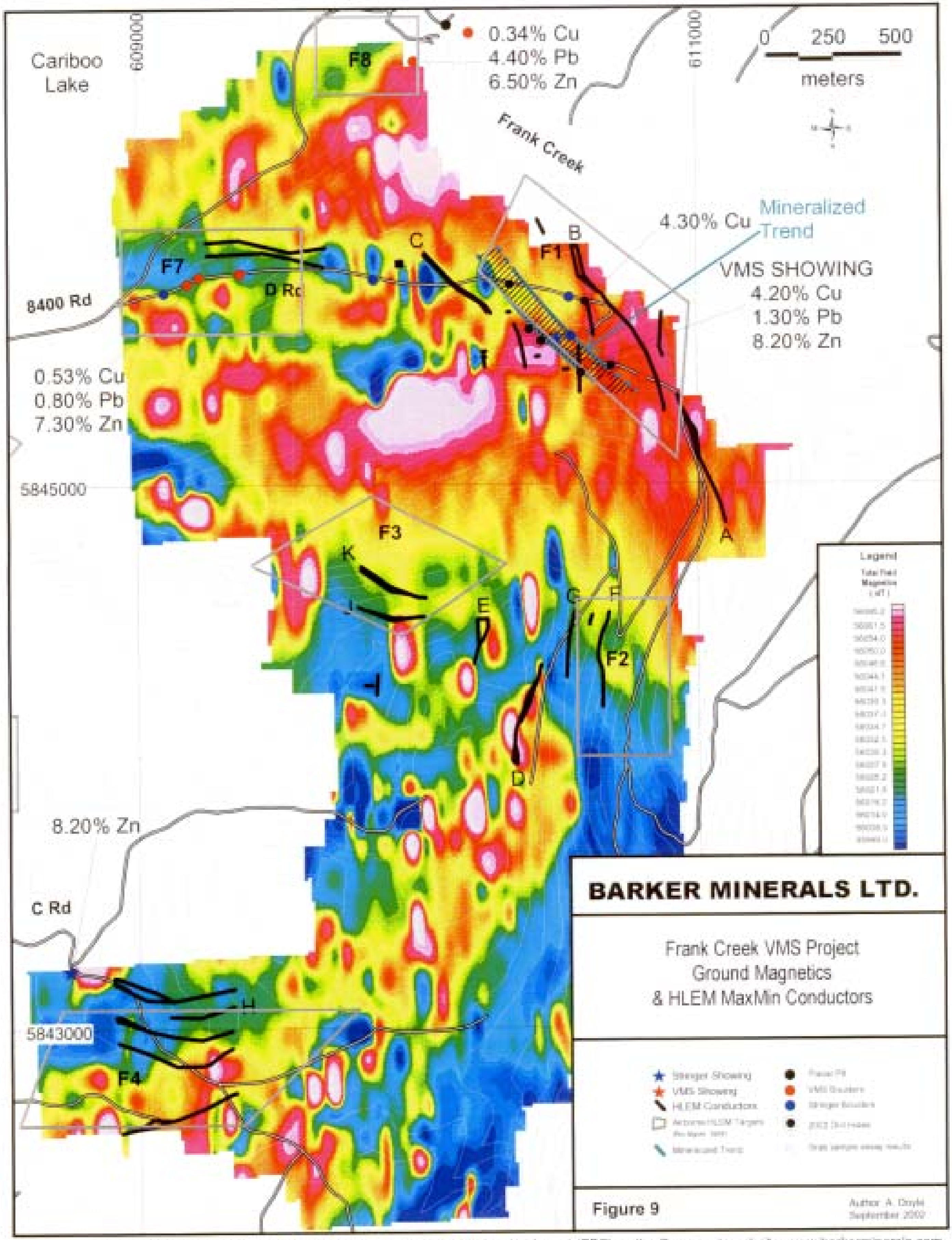

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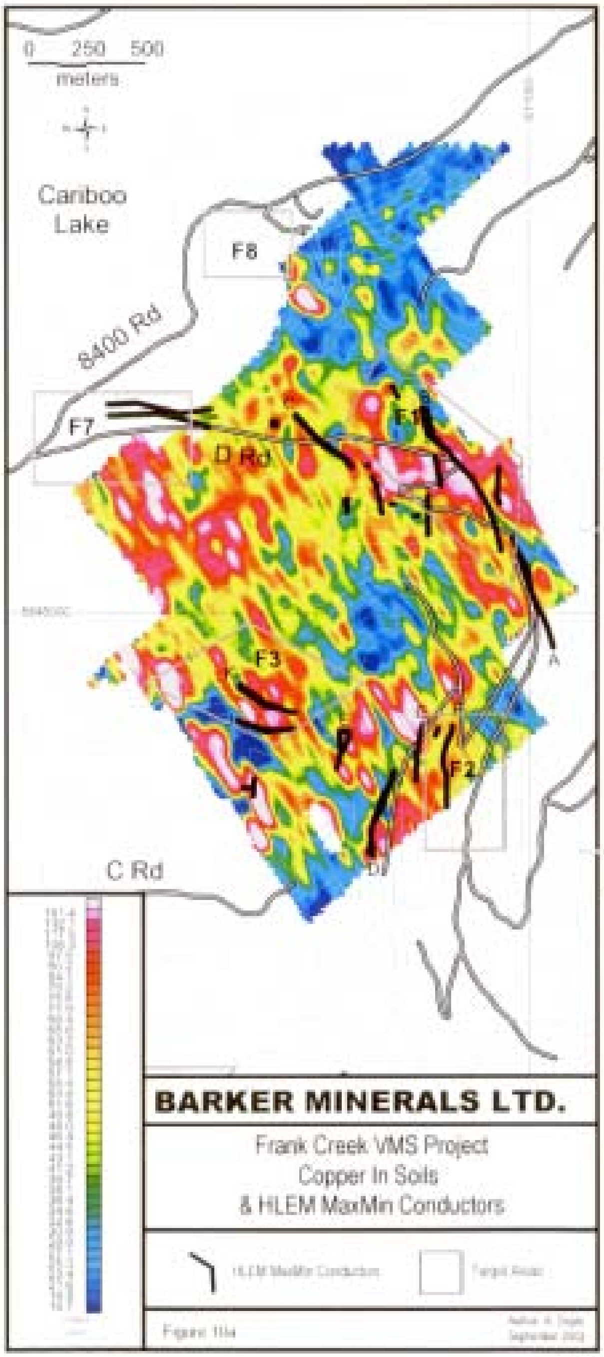

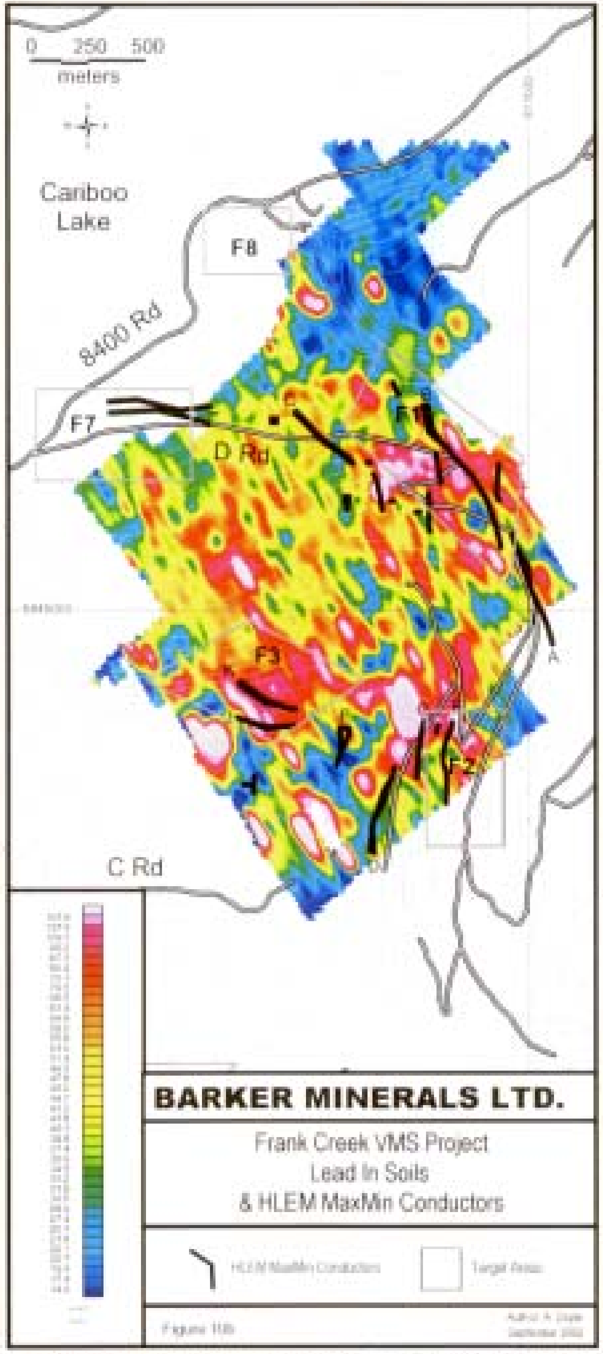

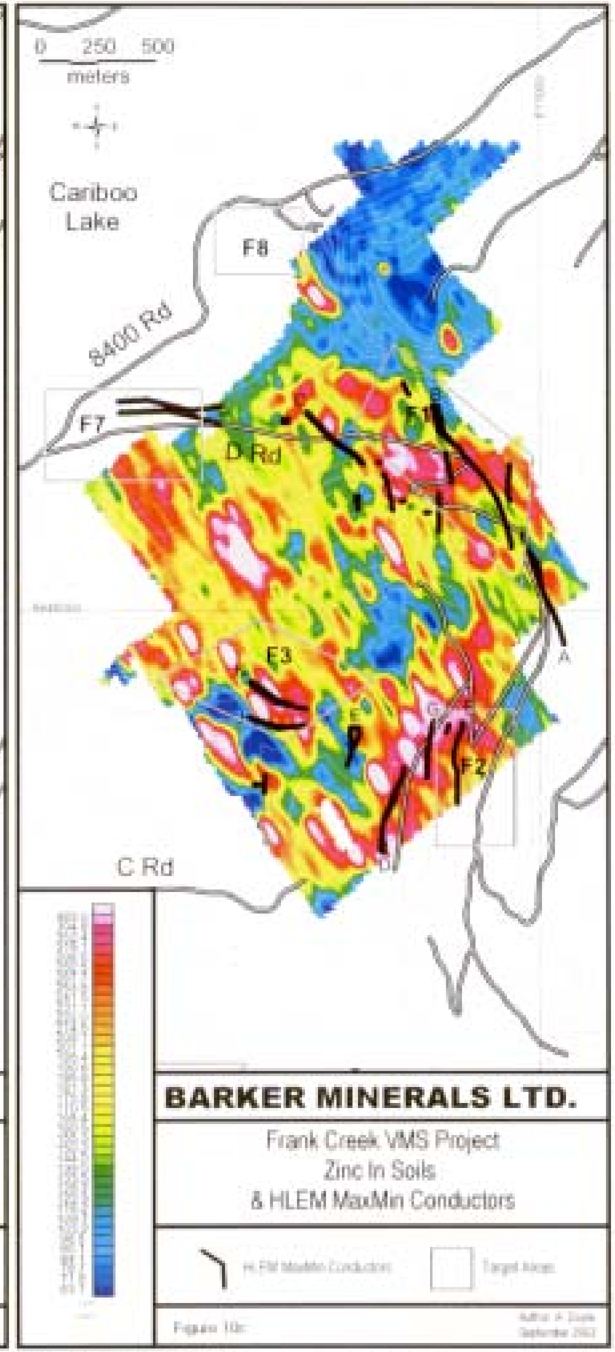

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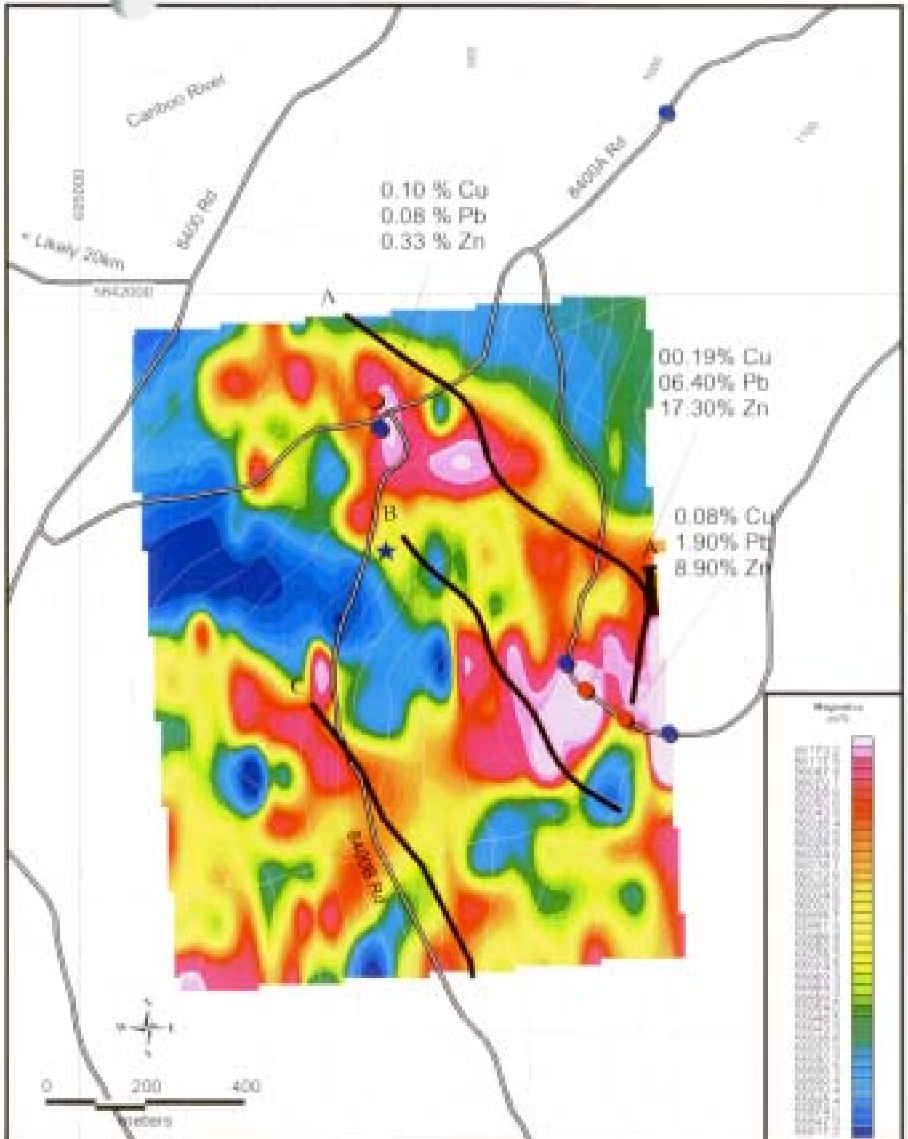
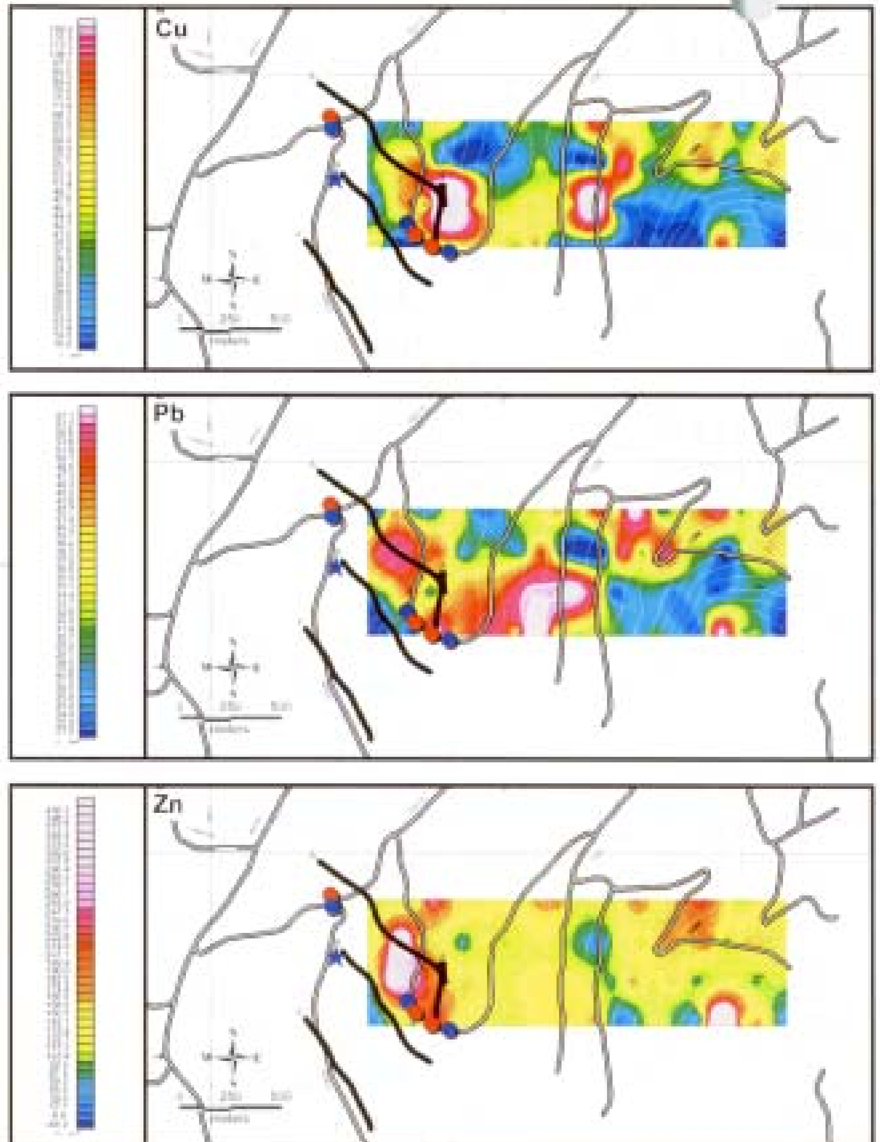


BARKER MINERALS LTD.

SCR VMS Project
Ground Magnetics
& HLEM MaxMin Conductors

Figure 11a



BARKER MINERALS LTD.

SCR VMS Project
Cu, Pb, and Zn in Soils (Casimiro Resources, 1986)
& HLEM MaxMin Conductors

Figure 11b

Stringer Showing
VMS Boulders
Stringer Boulders
HLEM Conductors
Grab sample assay results

* Enlargeable detailed figures are available for review in Adobe Acrobat format (PDF) on the Company's website. www.barkerminerals.com

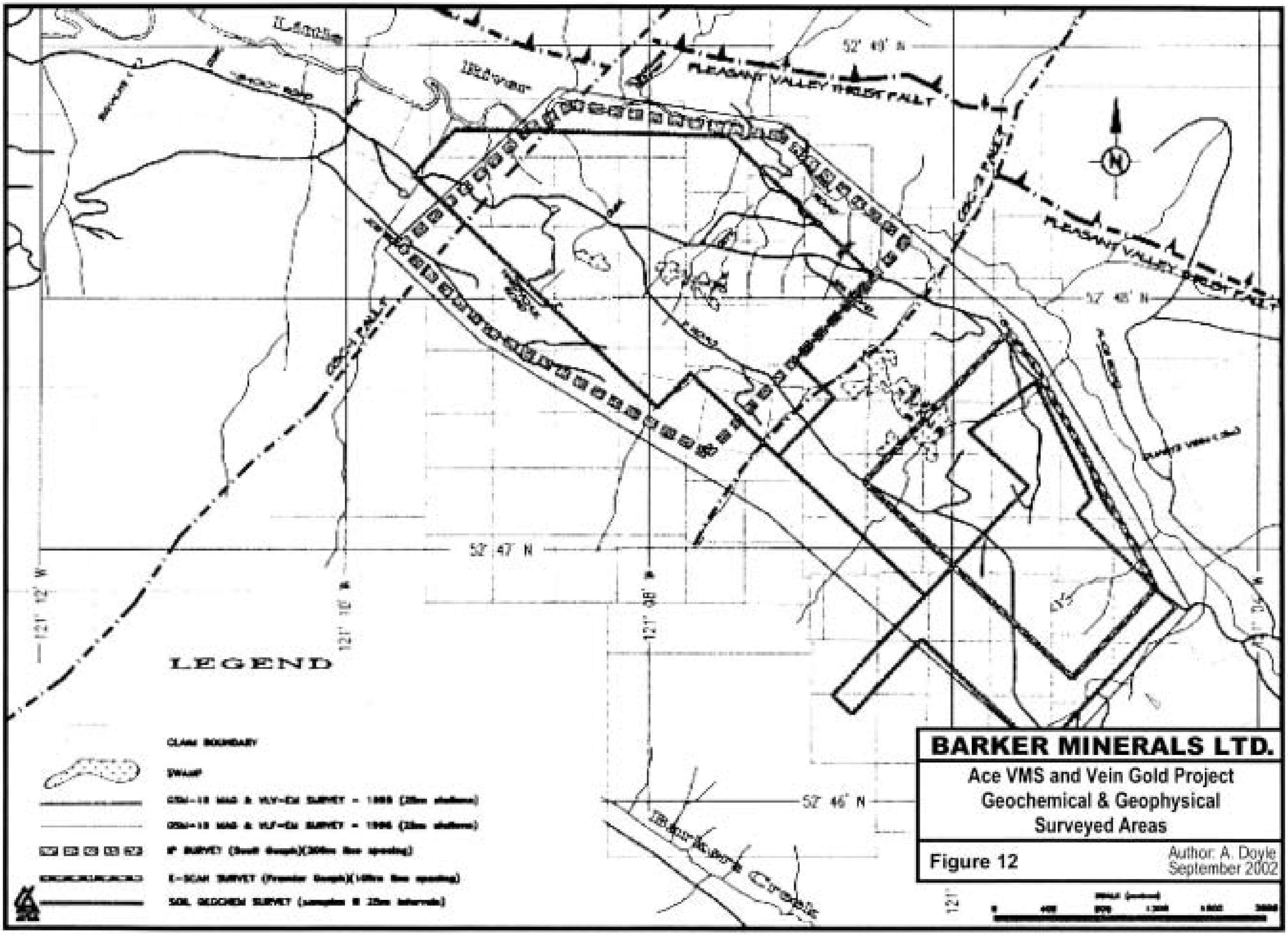


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Table 1 - Ace Project Assays

Sample #	Sample Type	Assay Type	Au ppm	Ag ppm	Cu ppm or %	Pb ppm or %	Zn ppm or %	As ppm	Sb ppm	Bi ppm	W ppm	Te ppm
1261	Q	2,3	18.8	10	.	.	.	.	.	2025	.	.
1199	Q	1,3	27.7	.	.	.	.	.	.	1965	.	.
CX-03	Q	1	18.2	6.8	.	.	.	.	.	1435	.	.
941160	Q	1	22.03	.	.	.	.	>10,000	.	35	.	.
1270	Q	3,1	13.3	.	.	.	.	.	.	386	.	.
1358	Q	1,3	23.7	.	.	.	.	.	.	510	.	46
1331	Q	3,1	2.5	.	.	.	.	.	.	250	337	.
951080	Q	3	7.6	.	.	.	.	.	.	775	224	.
1193	Q	3,1	8.2	.	.	.	.	.	.	318	781	28.6
1194	Q	3,1	5.4	.	.	.	.	.	.	289	210	26.2
9706	Q	3,4	6	.	.	.	.	213	3.1	1378	6	13.4
9709	Q	3,4	1.5	.	.	.	.	.	.	172	147	48
9798	Q	3,4	8.1	1.5	.	.	.	.	.	865	13	60.2

Massive Sulphide Boulders: Grab Samples

Sample #	Sample Type	Assay Type	Au ppm	Ag ppm	Cu ppm or %	Pb ppm or %	Zn ppm or %	As ppm	Sb ppm	Bi ppm	W ppm	Te ppm
295	MS	2,3	0.08	32.8	269	5.20%	8.50%	220	.	35	.	.
9752	MS	3	0.07	.	387	0.76%	7.2%	.	.	57	.	.
CX-45	MS	2	.	.	.	7.7%	8.8%	.	.	.	.	.
99AGN	MS	2	.	43	.	5.7%	9.9%	.	.	.	.	.

Cu-Stringer Boulders: Grab Samples

Sample #	Sample Type	Assay Type	Au ppm	Ag ppm	Cu ppm or %	Pb ppm or %	Zn ppm or %	As ppm	Sb ppm	Bi ppm	W ppm	Te ppm
1253	ST	3	0.25	27.2	4.1%	.	158	.	.	16	.	.

Legend:

Sample Type	Assay Type
MS - Massive sulphide boulders	**1** - Fire Assay
Q - Quartz sulphide vein boulders	**2** - Metal Assay
ST - Copper stringer boulders	**3** - ICP
	4 - Aqua Regia

Table 2. Frank Creek - Massive Sulphide and Cu-rich Stringer Zone Assays

Sample	Au ppb	Ag ppm	Cu %	Pb %	Zn %	Cd ppm	Fe %	S %	As ppm	Sb ppm	Bi ppm	Tl ppm	Se ppm	Mo ppm	Hg ppm	W ppm	Sn ppm
Massive Sulphide (Grab samples)																	
99-F1	**245**	**476**	0.62	**11.1**	**3.13**	88	44.7	30.8	1110	420	57	7	67	7	25	18	-
JD-49	**300**	**507**	0.29	**8.35**	1.10	30	-	-	2900	370	10	-	-	20	-	-	-
99-F2	**260**	**697**	0.30	**15.36**	**1.79**	60	45.9	33.9	6150	610	87	11	160	17	23	12	-
99-F3	**430**	**554**	0.61	**12.03**	**2.54**	79	45.0	33.5	8380	505	88	6	195	5	21	35	-
2306	44	**102**	0.28	**2.19**	**5.05**	172	30.6	18.8	346	31	285	4	80	12	25	7	14
JD-48	170	**47**	**4.47**	0.08	0.41	10	-	-	400	10	10	-	-	<10	-	-	-
99-F7	**600**	**102**	**1.37**	0.69	0.61	15	57.9	31.1	720	62	59	40	12	17	36	15	-
Cu-rich Stringer Zone (Grab samples)																	
99-F4	**290**	2.1	**7.41**	0.09	0.44	17	40.6	12.1	2810	4	42	<0.2	13	1	0.5	2	-
99-F5	**230**	1.3	**4.76**	0.07	0.36	13	41.1	11.4	2200	2	35	<0.2	12	1	0.4	3	-
Massive Sulphide Boulders from Frank Creek F-7 & F-8 Targets (Grab samples)																	
F-8 2029	188	**129**	0.34	**4.46**	**3.86**	121	50.7	40.1	4520	678	80	11	52	24	100	9	333
F-8 2030	173	**68**	0.33	**2.28**	**2.36**	52	45.5	35.1	28155	267	79	2	109	6	83	8	322
F-8 2031	80	**102**	0.16	**1.85**	0.52	18	34.8	28.1	3770	556	84	23	97	16	61	6	389
F-7 001-67	61	**65**	0.02	**0.53**	**7.3**	246	58	36	166	66	96	1	30	5.6	909	9	1000
F-1 Trench Bedrock (Grab samples)																	
F-1 Trench	131	**74**	**2.20%**	0.16%	323												
	150	**129**	**2.10%**	0.19%	0.13%												
	670	**104**	**4.50%**	0.24%	**0.66%**												
	45	12	0.28%	**0.99%**	**2.70%**												
	748	**115**	**3.10%**	**0.61%**	**2.03%**												
	704	**134**	**4.31%**	**0.51%**	**1.87%**												
TR-BW-10	5	4	0.42%	321	518												
	38	11	**0.88%**	538	951												
	32	1	**0.63%**	115	358												
	72	9	0.49%	**0.52%**	**2.42%**												
	109	14	0.42%	**1.46%**	**4.46%**												
Massive Sulphide Boulders from F-1 Target - Frank Creek (Grab samples by independent Majors)																	
226808	**490**	**204**	0.61	3.45	**8.21**	290	>30	>5	**778**	**88**	220	-	-	20	7	-	-
JD0008	-	**51**	2.23	. 43	0.75	-	28.4	-	-	-	-	-	-	-	-	-	-
JD0049	-	**506**	0.29	8.35	1.10	30	40		2900	**370**	-	-	-	-	-	-	-
14868	190	**344**	1.1	0.13	4.6	-	-	-	-	-	-	-	-	-	-	-	-

Massive Sulphide Chip Samples from F1 Target - Frank Creek (Chip samples by indepentdent Majors)

HD **.82%** Cu, .25% Zn, .21 % Pb, & **44 ppm** Ag, over 5.77 metres

209758 **2.1%** Cu over .77m, total width

14865 **2.1%** Cu, .28% Pb, .22% Zn, **134 ppm** Ag, 180 ppb Au, over 40cm

14869 **1.7%** Cu, .13% Pb, .57% Zn, **258 ppm** Ag, 190 ppb Au, over 60cm

Table 3. Frank Creek – Median Whole-Rock Analyses (Major Rock Types)

Unit	SiO2	Al2O3	Fe2O3	MgO	CaO	Na2O	K2O	TiO2	P2O5	MnO	Ba	Zr	Ti/Zr	*No.
Q	90.0	4.3	2.3	0.24	0.08	0.15	0.80	0.20	0.06	0.02	190	120	14.4	4
Fl	75.9	11.0	3.8	0.57	0.19	0.97	2.34	0.50	0.04	0.07	1310	200	13.5	13
QM	75.2	12.7	2.4	0.22	0.91	1.75	3.87	0.27	0.15	0.03	1490	72	22	10
P	60.3	17.3	7.0	2.24	0.12	0.85	4.10	0.50	0.09	0.06	1620	130	39	22
A	49.2	11.8	4.6	4.4	2.9	1.24	2.0	0.57	0.08	0.06	1520	66	56	7
Mf,t	45.3	14.4	9.9	6.6	6.3	2.35	1.42	2.20	0.27	0.14	2000	96	138	10
Md	40.3	10.8	10.7	7.6	8.2	1.6	0.41	1.38	0.16	0.17	152	60	148	4

Q quartzite
Fl felsic lapilli tuff or arkose
QM quartz monzonite (Quesnel Lake gneiss)
P phyllite (sericite and chlorite-rich
A argillite, shale
Mf,t mafic flow and tuff
Md Mafic dyke
*No. number of samples

Table 4. SCR – Semi-Massive Sulphide Mineralization

Sample	Au ppb	Ag ppm	Cu %	Pb %	Zn %	Cd ppm	Co ppm	Fe %	S %	As ppm	Sb ppm	Bi ppm	Se ppm	Mo ppm	Hg ppb	W ppm	Sn ppm
SCR-1	3	0.30	0.09	-	-	-	**148**	19.4	9.8	1	0.1	1.8	3.2	2.9	-	6	<1
SCR-3	8	3.40	0.03	0.05	0.02	0.10	95	20.9	4.2	1	0.1	16	15	2.1	-	1	**46**
SCR-6	**87**	0.15	0.09	0.02	-	0.10	**495**	40.2	24.9	78	2.5	**103**	**59**	3.4	30	6	2
SCR-7	*4*	*0.16*	*0.02*	*0.07*	*0.12*	*2.70*	*25*	*8.90*	*1.0*	*42*	*0.4*	3.2	*1.4*	**1.6**	*28*	**1**	*12*
SCR-7a	**34**	3.4	0.10	0.09	0.33	8.80	75	8.4	3.3	785	5.4	7.3	3.7	3.6	89	6	11
001-12	**60**	**33**	0.08	**1.80**	**8.90**	266	92	40	19	6	**11**	**60**	34	1.5	3380	1.7	3
001-14	**63**	**68**	0.01	**1.90**	**3.20**	123	24	18	9	1.1	1	**237**	25	1.2	1580	1.8	7
001-25	**80**	**12**	0.09	**1.20**	**3.80**	142	278	39	18	6.3	4	**24**	20	1.7	1740	2.7	4
2115	**168**	**56**	**0.19**	**6.40**	**17.30**	>500	-	26	4	7	28	**99**	91	6	7940	1	8

Table 5a.Ace VMS Project Drill Hole Parameters

Table 5a Ace VMS Project Drill Hole Parameters (1998 and 2002)

Hole No.	Easting (metres)	Northing (metres)	Elevation (metres)	Azimuth (metres)	Dip	Length (metres)
1	627094	5852254	963	180	-45	300.21
2	626510	5852284	1020	180	-45	199.63
3	624861	5852434	1022	180	-45	157.9
4	625101	5852337	1032	180	-45	199.94
5	627072	5851845	1073	180	-60	150.81
6	626184	5852090	1047	180	-70	134.11
7	625638	5852164	1045	180	-45	117.49
					Total Length	1260.09

2002 Drill Hole Parameters

Hole No.	Easting (metres)	Northing (metres)	Elevation (metres)	Azimuth (metres)	Dip	Length (metres)
Ace-02-01	6 + 65E	16 + 15S	1005	180	-45	144.78
Ace-02-02	8 + 64E	18 + 80S	1025	180	-45	164.90
Ace-02-03	5 + 70W	4 + 70N	1028	180	-45	150.88
Ace-02-04	5 +00W	4 + 20N	1025	180	-45	117.35
Ace-02-05	6 + 25W	5 + 00N	1027	n/a	-90	67.97
					Total	645.88

Table 5b - Ace VMS Project Summary Drill Logs (1998 and 2002)

1998

from (m) - to (m)			
98-01			
0 - 12.2	*overburden*		
12.2 - 130.4	quartz-rich schist, quartz-muscovite schist		
130.4 - 133.9	**felsite**		
133.9 - 136.8	argillite		
136.8 - 198.3	quartz-muscovite-chlorite schist		
198.3 - 209.7	quartz-calcite-muscovite-chlorite schist		
209.9 - 214.4	siliceous argillite		
214.4 - 270.9	quartz-muscovite-chlorite schist		
270.9 - 290.6	quartz-rich schist, minor quartz-muscovite schist		
290.6 - 300.2	quartz-muscovite schist		
from (m) to (m)			
98-02			
0 - 9.7	*overburden*		
9.7 - 27.2	quartz-muscovite-chlorite schist		
27.2 - 49.45	quartz-muscovite-biotite schist (biotite alteration)		
49.45 - 119.37	quartz-muscovite-chlorite schist, quartz-muscovite-biotite schist		
119.37 - 130.6	quartz-rich schist		
130.6 - 147.3	quartz-muscovite schist		
143.7 - 160.4	finely laminated carbonaceous meta-siltstone argillite		
160.4 - 176.55	quartz-muscovite-chlorite schist		
176.55 - 178.1	fault, abundant graphite		
178.1 - 181.45	quartz-chlorite-muscovite schist with quartz-tourmaline veins		
181.45 - 187.95	**recrystallized felsite**		
187.45 - 199.63	quartz-muscovite-chlorite		
from (m) to (m)			
98-03			
0.0 - 10.0	*overburden*		
10.0 - 24.75	quartz-muscovite		
24.75 - 106.25	**felsite**		
		39.25-39.5	**massive sulphide**
			calcite (probably limestone)
		39.8-40.0	**massive sulphide**
		106.25-110.72	quartz-muscovite-chlorite-schist
		110.72-119.6	siliceous argillite
		119.6-121.3	quartz-chlorite-muscovite schist
		121.3-127.4	**felsite**
		127.4-157.9	quartz-muscovite-chlorite schist (variable)
		98-04	
		0-6.9 m	*overburden*
		6.9-16.8	quartz-muscovite-(biotite) schist
		16.8-24.38	**felsite**
24.38 - 25.7	argillite, limestone (lost core)		
25.7 - 36.5	**felsite**		
36.5 - 39.5	mixed zone: quartz-chlorite-muscovite schist, felsite, limestone		
39.5 - 72.9	quartz-chlorite-muscovite schist		
72.9 - 79.1	quartz-muscovite-chlorite schist, minor felsite		
79.1 - 85.4	quartz-chlorite-muscovite schist		
85.4 - 89.55	**felsite**		
89.55 - 98.7	argillite, quartz-chlorite-muscovite schist		
98.7 - 101.8	**felsite**		
101.8 - 108.2	quartz-muscovite-chlorite schist		
108.2 - 109.3	**felsite**		
109.3 - 122.5	quartz-chlorite-muscovite schist		

122.5 - 199.94	quartzite, quartz-rich schist
from (m) to (m)	
98-05	
0 - 7.2	*overburden*
7.2 - 14.28	quartz-muscovite-chlorite-garnet schist
14.28 - 22.1	quartz-chlorite-muscovite schist
22.1 - 47.75	quartz-muscovite-chlorite schist, quartz-chlorite-muscovite schist
47.75 - 70.6	argillite, intermediate crystal tuff
70.6 - 113.57	quartz-rich schist, quartz-muscovite-chlorite schist
113.57 - 122.6	quartz-chlorite-muscovite schist
122.6 - 150.87	quartzite, quartz-rich schist
from (m) to (m)	
98-06	
0 - 4.57	*overburden*
4.57 - 7.92	argillite, calcareous argillite
7.92 - 18.1	**felsite**, minor calcareous argillite
18.1 - 34.53	quartz-muscovite-chlorite-garnet schist
34.53 - 39.1	argillite, cherty argillite, quartz-chlorite-muscovite schist, limestone
39.1 - 76.55	quartz-chlorite-muscovite schist
76.55 - 79	quartz-muscovite-chlorite schist
79.0 - 80.1	**felsite**
80.1 - 99.9	quartz-chlorite-muscovite schist, quartz-muscovite-chlorite schist
99.9 - 134.11	quartzite, quartz-rich schist, quartz-muscovite-chlorite schist
from (m) to (m)	
98-07	
0 - 13.34	*overburden*
13.34 - 27.7	quartz-muscovite-biotite schist (biotite alteration)
27.7 - 29.8	quartz-muscovite-chlorite schist
29.8 - 47.17	quartz-muscovite-biotite schist (biotite alteration)
47.17 - 91.88	**felsite**
	79.04-79.40 **semi-massive sulphide**
	91.88-112.3 quartz-chlorite-muscovite schist
	112.3-113.2 argillite
113.2 - 117.49	felsic to intermediate tuff

2002

DDH ACE-02-01

from (m) to (m)	
0.0 - 10.9	*overburden*
10.9 - 17.4	quartz-muscovite-biotite schist
17.4 - 19.0	quartz-muscovite-chlorite schist
19.0 - 22.0	quartz-chlorite-muscovite schist
22.0 - 26.6	quartz-muscovite-biotite schist
26.6 - 30.9	quartz-chlorite-muscovite schist
30.9 - 32.1	quartz-muscovite-chlorite schist
32.1 - 39.5	argillite
39.5 - 46.5	quartz-muscovite-chlorite schist
46.5 - 49.0	argillite
49.0 - 49.7	calc-silicate
49.7 - 50.7	marble
50.7 - 52.1	argillite
52.1 - 54.0	marble
54.0 - 55.2	quartz-muscovite-chlorite schist
55.2 - 58.8	argillite
58.8 - 65.3	**felsite**

65.3 - 67.5	quartz-chlorite-muscovite schist
67.5 - 71.2	quartz-muscovite-biotite schist
71.2 - 79.5	quartz-muscovite-chlorite-garnet schist
79.5 - 86.9	quartz-muscovite-chlorite schist
86.9 - 93.5	quartz-muscovite-chlorite-garnet schist
93.5 - 102.0	quartz-chlorite-muscovite schist
102.0 - 103.6	quartz-muscovite-chlorite-garnet schist
103.6 - 112.7	quartz-chlorite-muscovite schist (argillite)
112.7 - 132.8	quartz-muscovite-chlorite-garnet schist
132.8 - 135.0	argillite/quartz-graphite schist
135.0 - 144.8	quartz-muscovite-(chlorite) schist

DDH ACE-02-02

from (m) to (m)	
0.0 - 13.7	*overburden*
13.7 - 39.4	quartz-muscovite-chlorite-garnet schist
39.4 - 41.7	marble
41.7 - 46.8	quartz-muscovite-chlorite-garnet schist
46.8 - 47.7	quartz-muscovite-biotite schist
47.7 - 62.4	quartz-muscovite-chlorite schist
62.4 - 88.8	quartz-muscovite-chlorite-garnet schist
88.8 - 92.4	marble
92.4 - 105.9	quartz-muscovite-chlorite-garnet schist
105.9 - 108.5	fault
108.5 - 118.8	quartz-muscovite-chlorite schist
118.8 - 122.5	fault
122.5 - 164.9	quartz-muscovite-chlorite schist

DDH ACE-02-03

from (m) to (m)			
0.0 - 8.2	*overburden*		
8.2 - 55.5	**felsite**		
		8.5 - 9.7	**semi-massive sulphides (1.2m)**
		9.7 - 11.8	layered and disseminated sulphides
		11.8 - 50.3	felsite
		50.3 - 51.3	quartz-muscovite-chlorite schist
55.5 - 57.7	siliceous argillite		
57.7 - 59.6	**felsite**		
59.6 - 63.6	quartz-muscovite-chlorite schist		
63.6 - 70.8	**felsite**		
70.8 - 72.6	quartz-muscovite-chlorite schist		
72.6 - 78.9	**felsite**		
78.9 - 81.7	quartz-muscovite-chlorite schist		
81.7 - 95.1	argillite		
95.1 - 106.2	quartz-muscovite-chlorite schist		
106.2 - 107.7	marble		
107.7 - 113.9	quartz-muscovite-chlorite schist		
113.9 - 131.3	argillite		
131.3 - 150.9	quartz-muscovite-chlorite schist		

DDH ACE-02-04

from (m) to (m)

0.0 - 7.6	*overburden*			
7.6 - 25.3	quartz-muscovite-biotite schist			
25.3 - 28.1	quartz-muscovite-chlorite schist			
28.1 - 33.8	marble			
33.8 - 42.5	**felsite** (+marble)			
42.5 - 43.5	argillite/fault			
43.5 - 45.7	**felsite**			
45.7 - 46.2	argillite/fault			
46.2 - 53.8	**felsite**			
53.8 - 54.3	argillite/fault			
54.3 - 55.7	**felsite**			
		54.25- 54.6	**semi-massive sulphides (argillite)**	**(0.35m)**
		55.5 - 55.7	**semi-massive sulphides**	**(0.2m)**
				0.55m
55.7 - 56.6	argillite			
56.6 - 58.7	**felsite**			
58.7 - 60.3	argillite			
60.3 - 64.9	**felsite**			
64.9 - 65.9	marble			
65.9 - 67.2	**felsite**			
67.2 - 76.0	calc-silicate			
76.0 - 82.1	**felsite**			
82.1 - 82.9	marble			
82.9 - 84.1	**felsite**			
84.1 - 84.5	marble			
84.5 - 88.4	**felsite**			
88.4 - 91.1	quartz-muscovite-chlorite schist			
91.1 - 92.1	**felsite**			
92.1 - 94.5	argillite			
94.5 - 102.4	quartz-muscovite-chlorite schist			
102.4 - 104.3	**felsite**			
104.3 - 117.35	quartz-muscovite-chlorite schist			

DDH ACE-02-05

from (m) to (m)	
0.0 - 7.6	*overburden*
7.6 - 22.2	**felsite**
22.2 - 24.7	quartz-muscovite schist
24.7 - 29.6	**felsite**
29.6 - 36.9	quartz-muscovite schist
36.9 - 42.5	argillite
42.5 - 48.0	quartz-muscovite schist
48.0 - 51.3	argillite
51.3 - 54.5	**felsite**
54.5 - 56.4	argillite
56.4 - 57.7	**felsite**
57.7 - 63.3	argillite
63.3 - 63.8	fault
63.8 - 66.5	quartz-muscovite schist
66.5 - 67.97	argillite

Table 5c - Ace VMS Project Significant DDH intersections (1998 and 2002)

1998 Hole	Interval metres	Lth. (cm)	Au ppb	Ag ppb	Cu ppm	Pb ppm	Zn pmm	Mo ppm	Bi ppm	As ppm	Se ppm	Cd ppm	Sb ppm	Te ppm	Tl ppm	W ppm
98-03	**39.25-39.50**	**25**	**149**	**760**	**604**	**22**	**309**	**4.1**	**1.1**	**775**	**16.1**	**1.7**	**2**	*	*	*
	39.5-39.8	30	27	**5510**	85	56	**202**	**22**	*	136	4.3	1.3	1.2	*	*	*
	39.8-40.0	**20**	**280**	**1495**	**467**	**42**	**345**	**7.8**	**1.9**	**396**	**17.6**	**2.2**	**3.1**	*	*	*
	40.45-41.60	115	39	**556**	**242**	**57**	142	4.1	*	81	5.1		*	*	*	*
	41.6-42.67	107	15	486	102	**77**	**411**	**51**	*	59	4.7	4.8	*	*	*	*
	42.67-43.7	103	33	**673**	**247**	**77**	**519**	**79**	*	73	6.2	6.5	1.7	*	*	*
	43.7-45.5	180	9	**886**	**214**	**158**	**875**	4.2	*	26	6.9	5.5	*	*	*	*
98-04	27.4-30.9	150	**111**	485	177	**68**	**738**	**20**	2.3	194	5.9	5.8	*	*	*	*
98-06	9.7-11.0	130	3	450	195	45	**267**	**25**	2.2	*	7.7	1.8	*	*	*	*
	11.0-12.04	104	2	**645**	79	**64**	104	**16.5**	3.1	*	3.9	1.1	*	*	*	*
98-07	**79.04-79.4**	**36**	**6**	**1195**	**538**	**36**	**127**	**11**	**15.2**	*	**15.9**		*	*	*	*
	84.5-86.0	150	5	**2790**	44	**376**	**460**	**75**	9.1	38	4.5	11.2	*	*	*	*

2002 Hole	Interval metres	Lth. (cm)	Au ppb	Ag ppb	Cu ppm	Pb ppm	Zn pmm	Mo ppm	Bi ppm	As ppm	Se ppm	Cd ppm	Sb ppm	Te ppm	Tl ppm	W ppm
02-01	13.65-13.80	15	8	380	**285**	**342**	14	6	**626**	0.2	1	0.16	**0.25**	**0.95**	0.24	**77**
	71.56-71.63	7	**764**	**1300**	**339**	**70**	106	4	**26.9**	0.2	3	0.06	**0.25**	**0.55**	**1.02**	**23**
02-02	159.63-159.75	7.5	**692**	400	72	30	40	6	**72**	0.2	*	0.02	0.05	**0.25**	0.3	**18**
02-03	21.0-21.6	60	*	**760**	35	**92**	**844**	**116**	4.7	0.2	3	16.6	0.05	0.05	0.18	8
	21.6-21.8	20	*	**540**	65	47	**628**	**26**	4.8	0.8	5	3	**0.25**	0.05	**0.72**	**12**
	21.8-22.35	55	*	340	78	45	**502**	**102**	1.5	0.2	5	6.3	**5**	0.1	0.22	7
	22.35-23.35	100	*	**700**	72	**73**	**446**	**104**	**10.3**	0.2	5	5.6	0.1	0.1	0.18	6
02-04	42.5-43.5	100	36	280	27	**63**	**278**	15	1.7	**35**	5	2.4	1.4	0.15	**0.92**	3
	43.5-43.75	25	**526**	360	113	32	70	8	2.5	**318**	7	0.6	**3.6**	**0.25**	0.5	5
	45.7-46.2	50	16	180	127	8	**732**	**28**	0.6	**23**	10	7.5	**1.1**	0.05	**1.26**	4
	54.25-54.6	35	34	**1440**	**663**	17	**234**	14	2.02	**42**	23	0.4	**2.6**	**1**	**0.72**	3
	57.6-57.9	30	14	**1420**	**486**	**50**	28	**42**	**29.4**	5	15	0.3	**0.25**	**0.25**	**0.28**	**12**
	58.5-58.67	0.15	42	**1640**	**571**	**68**	**206**	**84**	**31.5**	**69**	13	2.2	**0.45**	**0.25**	0.22	9

Table 6a - Frank Creek VMS Project Drill Hole Parameters (2002)

Hole No.	Date Collared	Date Completed	Northing (metres)	Easting (metres)	Elevation (metres)	Azimuth (metres)	Dip	Length (metres)
FC-02-01	15-May-02	16-May-02	57 + 95 N	16 + 00 W	1042	90	-45	97.84
FC-02-02	16-May-02	18-May-02	57 + 20 N	12 + 80 W	1077	90	-45	144.78
FC-02-03	18-May-02	20-May-02	54 + 30 N	13 + 47 W	1140	90	-45	118.87
FC-02-04	20-May-02	22-May-02	54 + 50 N	12 + 20 W	1140	90	-45	158.19
FC-02-05	22-May-02	24-May-02	55 + 60 N	15 + 00 W	1115	75	-45	158.5
FC-02-06	24-May-02	25-May-02	56 + 11 N	15 + 41 W	1103	70	-45	135.03
							Total	**813.21**

Table 6b - Frank Creek Project: Summary Drill Hole Core Logs (2002)

DDH FC-02-01

from (m) to (m)				
0.0 - 6.1	*overburden*			
6.1 - 11.2	quartz-sericite phyllite - breccia			
11.2 - 15.9	argillite			
15.9 - 18.2	quartz-sericite phyllite - breccia			
18.2 - 20.8	argillite			
20.8 - 24.9	quartz-sericite phyllite			
24.9 - 26.9	argillite			
26.9 - 29.8	quartz-sericite phyllite			
		26.9 - 27.2	**disseminated sulphide zone (10-15%)**	**(0.3m)**
		29.1 - 29.5	**disseminated sulphide zone (10-15%)**	**(0.4m)**
				(0.7m)
29.8 - 30.8	fault			
30.8 - 47.5	argillite			
47.5 - 49	fault			
49.0 - 51.3	quartz-sericite phyllite			
51.3 - 59.4	argillite			
59.4 - 60.1	fault			
60.1 - 68.2	quartz-sericite phyllite			
68.2 - 71.3	fault			
71.3 - 74.7	quartz-sericite phyllite			
74.7 - 77.2	fault			
77.2 - 81.5	quartz-sericite phyllite			
81.5 - 82.7	fault			
82.7 - 87.2	quartz-sericite phyllite			
87.2 - 97.84	quartz-sericite-chlorite phyllite			

DDH FC-02-02

from (m) to (m)	
0.0 - 10.7	*overburden*
10.7 - 28.2	quartz-sericite phyllite
28.2 - 32.2	argillite
32.2 - 36.3	quartz-sericite-chlorite phyllite
36.3 - 38.7	quartz vein / fault
38.7 - 41.2	fault
41.2 - 48.7	quartz-graphite schist
48.7 - 51.5	quartz-sericite phyllite - breccia
51.5 - 53.5	fault
53.5 - 61.0	quartz-sericite phyllite
61.0 - 63.8	fault
63.8 - 70.6	quartz-sericite phyllite
70.6 - 75.6	argillite
75.6 - 76.0	fault
76.0 - 84.1	argillite
84.1 - 95.2	quartz-sericite phyllite - breccia
95.2 - 101.0	quartz-sericite phyllite - grit
101.0 - 114.3	quartz-sericite phyllite - breccia
114.3 - 120.9	fault
120.9 - 144.78	quartz-sericite phyllite - breccia

DDH FC-02-03

from (m) to (m)

0.0 - 5.2	*overburden*
5.2 - 14.1	quartz-sericite phyllite
14.1 - 14.4	quartz vein
14.4 - 15.9	fault
15.9 - 24.2	quartz-sericite-chlorite phyllite
24.2 - 41.1	fault
	34.2 - 34.7 massive sulphides (0.5m)
41.1 - 41.8	andesite dyke
41.8 - 42.5	fault
42.5 - 43.7	quartz-sericite phyllite - grit
43.7 - 44.2	argillite
44.2 - 44.4	fault
44.4 - 45.1	andesite dyke
45.1 - 50.1	quartz-sericite phyllite
50.1 - 50.2	fault
50.2 - 53.4	argillite
53.4 - 54.4	quartz-sericite phyllite - grit
54.4 - 56.2	argillite
56.2 - 66.5	quartz-sericite phyllite - grit
66.5 - 68.0	quartz-chlorite-sericite phyllite
68.0 - 70.5	fault
70.5 - 77.7	quartz-chlorite-sericite phyllite
77.7 - 79.2	quartz-sericite phyllite
79.2 - 79.7	quartz-chlorite-sericite phyllite
79.7 - 80.9	fault
80.9 - 90.0	quartz-sericite phyllite - grit
90.0 - 94.6	argillite
94.6 - 96.1	quartz-sericite phyllite - grit
96.1 - 101.8	fault
101.8 - 103.4	quartz-sericite phyllite - grit
103.4 - 107.0	fault
107.0 - 111.7	quartz-sericite phyllite
111.7 - 113.9	fault
113.9 - 118.87	quartz-sericite (chlorite) phyllite

DDH FC-02-04

from (m) to (m)	
0.0 - 5.2	*overburden*
5.2 - 13.6	quartz-chlorite-sericite phyllite
13.6 - 17.1	quartz sericite phyllite
17.1 - 43.6	quartz-chlorite-sericite phyllite
43.6 - 46.0	quartz sericite phyllite
46.0 - 47.4	argillite
47.4 - 50.5	quartz-sericite phyllite
50.5 - 59.0	limestone breccia
59.0 - 64.1	quartz-sericite phyllite - breccia
64.1 - 64.9	fault
64.9 - 65.8	quartz-sericite phyllite - breccia
65.8 - 67.9	argillite
67.9 - 69.3	argillite - breccia
69.3 - 70.0	quartz-sericite phyllite
70.0 - 71.2	argillite - breccia
71.2 - 71.9	fault
71.9 - 73.3	argillite - breccia
73.3 - 76.7	limestone breccia

76.7 - 86.5	quartz-sericite phyllite
86.5 - 86.9	fault
86.9 - 107.5	argillite
107.5 - 108.7	fault
108.7 - 114.3	quartz-chlorite-sericite phyllite
114.3 - 127.8	fault
127.8 - 130.2	quartz-sericite phyllite
130.2 - 144.6	quartz-chlorite-sericite phyllite
144.6 - 147.7	quartz-sericite phyllite
147.7 - 158.19	quartz-chlorite-sericite phyllite

DDH FC-02-05

from (m) to (m)

0.0 - 5.1	*overburden*			
5.1 - 8.9	argillite			
8.9 - 14.5	quartz sericite phyllite			
14.5 - 21	argillite			
21.0 - 32.9	quartz sericite phyllite (5-10% disseminated sulphides)			
32.9 - 37.2	argillite			
37.2 - 139.9	quartz sericite phyllite			
		50.3 - 55.6	**disseminated sulphide zone (10-15%)**	**(0.5m)**
		55.6 - 56.3	**semi-massive sulphide zone (25-40%)**	**(0.7m)**
		56.3 - 57.9	**disseminated sulphide zone (10-15%)**	**(1.6m)**
				(2.8m)
139.9 - 151.9	fault			
151.9 - 158.50	quartz-sericite phyllite - grit			

DDH FC-02-06

from (m) to (m)

0.0 - 5.2	*overburden*			
5.2 - 10.0	argillite			
10.0 - 15.7	quartz sericite phyllite			
15.7 - 16.9	argillite			
16.9 - 27.2	quartz sericite phyllite			
27.2 - 27.8	fault			
27.8 - 33.4	quartz sericite phyllite			
33.4 - 33.9	fault			
33.9 - 54.0	quartz sericite phyllite (bands of disseminated sulphides)			
54.0 - 54.8	fault			
54.8 - 56.4	quartz sericite phyllite (bands of disseminated sulphides)			
56.4 - 56.7	fault			
56.7 - 135.03	quartz sericite phyllite			
		57.6 - 58.5	**disseminated sulphide zone (10-15%)**	**(00.9m)**
		58.5 - 58.6	**semi-massive sulphide zone (25-40%)**	**(00.1m)**
		56.3 - 75.2	**disseminated sulphide zone (5-10%)**	**(18.9m)**
				(19.9m)

Table 6c - Frank Creek VMS Project Significant DDH Intersections (2002)

	From	To	Length	Au ppb	Ag ppm or oz/t	Cu ppm or %	Pb ppm or %	Zn ppm or %
FC-02-01	22	30.8	8.8	*	2.7	222	696	0.40%
	26.9	29.5	2.6	*	5.8	**560**	**765**	**1.10%**
incl	26.9	27.3	0.4	*	18	**0.11%**	**0.30%**	**2.30%**
incl	28.96	29.1	0.14	14	11.5	**0.25%**	**0.10%**	**3.42%**
and	29.1	29.5	0.4	*	12.4	**0.14%**	**808**	**3.40%**
FC-02-03	34.2	34.7	0.5	*	**2.6 oz/t**	**0.52%**	**0.28%**	**0.33%**
FC-02-05	6	6.3	0.3	114	15.4	**1.14%**	**486**	**0.10%**
	19.5	25.7	6.2	*	5.2	0.19%	407	0.15%
incl	20.2	20.3	0.1	726	15.7	**0.90%**	**289**	**0.41%**
	22.8	26.5	3.7	48	4.2	0.12%	400	0.11%
	29.5	30.4	0.9	174	14.9	**0.72%**	**578**	**0.24%**
	38.7	38.9	0.2	38	24.6	**0.40%**	**0.18%**	**0.26%**
	48.7	59.4	10.7	*	2.7	0.14%	168	0.13%
incl	51.8	56.3	4.5	*	3.4	0.24%	232	0.24%
incl	54.8	56.3	1.5	*	4.9	**0.34%**	**264**	**0.39%**
FC-02-06	17.2	17.5	0.3	14	4.2	0.15%	324	618
	27.2	27.28	0.08	216	11.5	**302**	**0.14%**	**0.51%**
	56.7	59.4	2.7	*	8.1	0.15%	0.10%	0.23%
incl	58.5	58.95	0.45	*	**2.6 oz/t**	**2.08%**	**0.54%**	**0.98%**
	62.14	62.2	0.06	190	8.7	**0.34%**	**298**	**0.79%**
	62.6	63.09	0.49	22	4.5	0.18%	201	618
	64.9	69	4.1	*	3.2	0.17%	68	270
incl	67	69	2	60	2.4	0.15%	90	236
incl	67.75	67.8	0.05	584	15.3	**1.27%**	**325**	**842**
incl	68.2	68.35	0.15	462	13.4	**0.78%**	**219**	**0.12%**
	86.6	86.8	0.2	132	2.2	0.16%	70	380

Table 7 - Quesnel Platinum Project Rock Samples and Assays

Sample #	Project name	Cu (ppm)	Pb (ppm)	Zn (ppm)	Ag (ppb)	Au (ppb)	Pt (ppb)	Pd (ppb)	Rh (ppm)	Ni (ppm)	Co (ppm)	Cr (ppm)	MgO (%)	Bi (ppm)	As (ppm)
20/05/02	Mag	44.5	2.07	44.6	51	0.9	<3	<3		22.8	36	5.4	7.45	0.05	0.9
2001	Mag	10.3	2.09	53.3	20	1	<1	<1		4.5	15.8	4	7.9	0.03	1
2008	Mag	11.3	2.2	56.8	32	<1	1	<1		13.4	37	7.6	8.5	0.04	1.3
2021	Mag	305	5.22	68.8	233	9	3	2		60.9	40.3	140	3.3	0.23	20.3
2026	Mag	178	3.07	25	82	20	4	2		26.3	42	160	7.6	<.02	3.2
2050	Gerimi Mag	13.90%	9.75	93	38000	90	8	55		6.9	11.7	22.2	1.6	0.04	24
2075	Gerimi	3498	7.4	82.5	796	4	5	6	<1	4.6	14.3	6.4	3.2	<.02	21
2076	Gerimi	55206	10.45	52.1	22087	6	9	18	3	48.4	25.3	136.9	4.7	<.02	50
2077	Gerimi	7206	5.11	93	3266	1	9	7	3	24.2	29.1	24.9	8.6	<.02	5.1
2081	Gerimi	1.5%	11.66	97.9	6309	8.7	4	13	<1	13.6	19.3	18.4	4.1	0.04	14
2083	Gerimi	131	10.78	67	72	1.8	3	11	5	126	34	310	7.3	0.04	18
2002	20 Mile	92	8.47	95.6	226	1	3	5		19.4	19.7	29.6	3.5	0.12	4.8
2003	20 Mile	50	1.38	93.7	37	1.2	1	<1		204.7	51.3	470	10.9	<.02	0.4
2045	20 Mile	54	2.43	97.9	55	<1	<1	1		234	49	520	10.5	<.02	0.2
2089	20 Mile	313	6.15	64.4	198	4	<2	10	<5	27	32	90	5.4	0.06	0.7
2090	20 Mile	65	6.15	69	53	<2	3	9	<5	85	33	390	5.7	0.04	2.3
2091	20 Mile	108	6.02	77.8	55	3	<2	7	<5	54	27	260	4.8	0.09	3.3
2092	20 Mile	96	1.35	75.7	36	<2	<2	5	<5	38	31	120	5.3	<.02	0.4
2093	20 Mile	174	2.46	55.8	70	4	<2	7	<5	22	25	70	4.2	0.02	3.3
2094	20 Mile	73	4.09	56.3	86	13	4	10	6	42	16	190	2.7	0.06	16
2095	20 Mile	154	3.76	59.2	53	16	3	3	<5	29	19	80	3.5	<.02	0.7
2096	20 Mile	237	2.33	55.6	554	<2	7	9	<5	58	36	240	5.8	<.02	1.8
BSA-1	20 Mile concentrate					51018	891	16	10						
BSA-2	20 Mile concentrate					99999	4832	118	167						
20/06/10	20 Mile/ Maude	60	6.01	81.8	64	3	7	8		16.2	18.4	130	4.9	0.05	2.8
2025	Maude	91	7.42	73	159	1	4	2		37.8	15.9	110	3.7	0.12	9.6
2025-A	Maude	70	5.76	101	97	1	<1	2		20.9	23.4	37.5	5.1	0.06	12
2025-B	Maude	68	6.63	83.1	100	5	1	2		16.7	16.7	14.6	2.2	0.08	10
2040	Maude	147	18.67	111	169	10	4	10		9.4	24	13.3	4.7	0.08	1.7
2041	Maude	295	22	133	344	6	<1	3		29.8	23.3	55.3	8.8	0.05	9
2041-A	Maude	23.67	7.71	66.4	57	2	8	10		37.1	30.1	500	8.83	0.06	6.7
2042	Maude	127	20	118	136	8	1	1		5.9	16	6.6	2.2	0.13	7.9

2043	Maude	3181	13	80.7	2292	10	3	3		13	17.6	21.9	3.4	0.12	19	0.34	519	56.9	0.188	42	6.9
2043-A	Maude	89	6.55	82.1	71	1	16	14		21.5	30.4	250	8.7	0.02	16	1.1	6	35.2	0.237	45	11.5
2044	Maude	231	8.27	95.3	128	4	2	15		7.4	29	3.8	4.4	0.15	3.6	0.13	3	67.3	0.269	47	10.22
2097	Maude	129	2.07	81	76	2	9	11	<5	38	33	170	6.9	0.04	0.3	0.1	14	72	0.269	47	10.2
2098	Maude	85	2.95	82	62	3	4	7	<5	79	38	120	6.9	<.02	0.9	0.1	9	70	1.3	44	13.6
2100	Maude	206	5.41	73	97	3	4	13	<5	25	33	90	4.8	0.1	2.3	0.11	98	65	0.215	46	10.3
2004	Coyote Porter	62	9.37	69	148	2	3	3		14.4	12.9	10.8	1.9	0.09	3.5	0.45	12	117	0.6	65	5
2005	Coyote Porter	132	5.46	47.2	66	<1	11	14		49.5	41	150	4.9	0.1	13	0.33	17	80	0.8	56	7
2047	Coyote Porter	92	8.27	81.6	110	5	<1	5		15.9	22.1	30	3.7	0.07	4.9	0.28	12	63	0.353	53	8.2
2060	Coyote Porter	71	11.04	48	263	2	1	4	<1	22.7	12.5	15.6	1.6	0.8	4.3	0.37	32	111	0.269	69	4
2061	Coyote Porter	120	11.58	220	488	5	5	6	<1	27.3	22.5	22.9	3	0.62	12	1.9	51	90	0.074	57	9
2062	Coyote Porter	111	4.79	81	96	1	5	5	1	20.5	30.9	40.2	3.8	0.56	6.8	0.11	1	77	0.354	54	9
2006	Porter	29.37	1.57	13.7	20	<1	3	<1		1587.6	87.5	2240	33	0.04	5.8	0.18	5	3	0.004	43	9.6
2007	Porter	13.86	12.62	46.3	71	1	<1	<1		27.1	6.5	18.6	1.2	0.06	6.4	0.28	30	376	0.5	83	2.7
2024	Porter	6.85	1.89	14.4	20	<1	6	3		1,900	93.8	3780	38	0.04	9.3	0.28	<5	3	0.004	38	4.53
2024-A	Porter	10.84	5.28	11.7	34	2	9	7		1882.8	82	2590	41	0.05	11.8	0.2	20	3.3	0.001	39	3.58
2079	Porter	37.02	0.67	51.5	40.5	1	5	1	8	32	23	26.7	5.6	<.02	1.1	0.1	<5	67	1.5	54	11.2
2080	Porter	14.47	2.02	10.7	23	2	1	1	<1	1865	99	3590	37.1	0.06	5.6	0.07	20	2.5	0.004	42	6.9
2010	Victoria	128	6.16	54.9	59	3	8	7		26.9	34	35.1	6.2	0.07	1.7	0.13	19	50	0.9	49	10.1
2028	Victoria	139	7.28	70.9	74	1	11	8		33.3	28.1	51.6	6.5	0.06	3.1	0.14	13	54.8	0.279	47	10.9
2013	Kangaroo	8.54	5.26	19.9	81	3	2	<1		1214.3	69.4	858.9	32	0.05	20	1.2	<5	7.4	0.019	36	6.1
2014	Kangaroo	77	4.42	60.4	53	35	1	<1		14.7	16.9	50	2.8	0.03	5.9	0.36	14	91	0.276	58	7.3
2015	Kangaroo	154	5.91	59.7	134	4	2	2		25.1	28.2	90	3.7	0.14	0.1	0.29	<5	69	0.149	53	8.8
2015-A	Kangaroo	152	11.87	65.9	176	3	2	6		39.5	19.8	170	2.8	0.36	<.1	0.22	5	87	0.124	56	7.2
2015-B	Kangaroo	13.63	0.69	59.2	16	2	11	6		178	29.4	1109	17.6	0.02	7.4	0.21	<5	21	0.094	44	10
2016	Kangaroo	126	5.9	178	266	4	3	5		37	18.7	100	3.2	0.14	0.9	0.63	33	93	0.095	59	6.3
2018	Kangaroo	150	4.49	105	167	1	2	3		44.5	24.7	190	4.2	0.4	1.9	0.2	<5	69.9	0.128	50	7.9
2019	Kangaroo	146	3.65	67	52	1	5	9		73	36	210	6.6	0.14	0.4	0.13	6	44.3	0.122	49	10.6
2020	Kangaroo	509	3.09	54.5	302	5	11	12		69.7	64.2	190	7.7	0.14	0.9	0.16	5	30.8	0.093	45	11.4

2046	Kangaroo	51.34	4.38	82	62	2	<1	3		10.3	22	12.3	2.7	0.06	4.5	0.3	5	75.6	0.25	58	5.9
2082	Kangaroo	93	2.97	5.7	128	3	4	4	4	814	54	1960	16	0.1	151	91	170	14.8	<.001	39	4.3
2084	Rollie	5.05	4.14	8.6	20	21	4	5	4	1185	77	2500	28.7	0.12	17	0.12	24	2.1	0.001	30	7.2
2085	Rollie	4.33	0.55	9	27	28	6	3	2	1409	87	2830	29.3	0.13	1.1	0.13	9	0.5	0.004	28	5.6
2086	Rollie	38	3.28	94	258	4	2	5	5	38	37	60	5.3	0.06	<.1	0.06	17	76.4	0.031	47	11.4
2115	Rollie	3.89	31.48	40.5	78	8	3	6	<5	1309	78	2400	25.6	0.04	<.1	0.18	<5	14.6	0.001	31	6.8
2116	Rollie	5.65	10.31	24.3	18	<2	5	3	<5	19.3	43	580	10.5	<.02	<.1	0.07	<5	23.2	0.05	50	8
2117	Rollie	17.28	9.34	50.7	36	<2	<2	3	<5	81.5	30	200	6.1	<.02	0.1	0.19	<5	119.4	0.195	50	9.7
2023	Nyland	36.5	3.44	56.5	50	3	<1	<1		67	22.7	190	4.2	0.03	2.2	0.03	<5	42.2	0.057	53	8.5
2311	Nyland	45	2.12	40.2	42	<2	2	3	<5	27	34	150	6.1	<.02	<.1	0.07	10	35.9	0.145	50	10.8
2312	Nyland	44	1.93	48.6	37	4	4	3	<5	10.3	26	23.4	4.3	<.02	0.1	0.07	<5	49.2	0.133	52	9.8
2313	Nyland	17	9.05	68.8	101	2	<2	<2	<5	27	7.7	70.8	2.3	0.5	0.3	0.14	5	107.3	0.123	66	3.4
2314	Nyland	4.41	6.14	22.2	31	17	<2	<2	<5	8.9	8.4	16.7	1.1	0.09	<.1	0.14	6	45.7	0.082	70	4
2315	Nyland	75	2.92	41.1	77	4	<2	3	<5	144	46	650	9.8	0.03	0.9	0.12	9	31.4	0.131	48	10.5
2316	Nyland	22	1.61	33.8	44	<2	3	5	9	9	13	13.3	2.6	<.02	1.2	0.1	<5	75.6	0.099	59	6.8
2317	Nyland	13	1.37	44.5	29	2	<2	3	<5	23.3	30.7	56.3	3.7	0.04	0.9	0.15	<5	132.2	0.085	53	8.6
2078	BB	74	6.81	78	59	<1	<1	<1	2	25.6	121	128	0.29	0.31	19	0.21	<5	173.3	<.001	66	6.7

Table 8. Analytical Check Samples

Lab	Sample #		Sample #	A/T		Au ppb		Cu %	Pb%		Zn %		Te ppm	Sample type
Eco-tech	A		O2-95	3					5.2		8.5			
Acme														Ace / VMS
Chemex	C	B	21	3					7.O	9.O	7.7	8.8		
Eco-tech	A		1199	2		27,710								
Acme	C	F	1199	2		28,972	27,341							Ace/Au/Qtz
Eco-tech	A		1358	2		23,710								
Acme														Ace/Au/Qtz
Chemex	D	E		2		17,310	16,590							
Acme	A		95-1273	2		8,028								Ace/Au/Qtz
Chemex	C		O4	2		8,000								
Acme	A		9769	1									65.8	Ace / Te
Chemex	C		O5	3									36.O	
Acme	A		97101	1									37.O	Ace/ Te
Chemex	C		O9	3									35.O	
Acme	A	D	1195	2	2A	2,995	.102/opt							Ace/Au/Qtz
Acme	A	D	1193	2	2A	8,258	.238/opt							Ace/Au/Qtz
Acme	A		9769	1		612								Ace/Au/Qtz
Chemex	C		O5	2		270								
Acme	A	D	99-F-1	3	3				11.1	11.2				Frank/VMS
Eco-tech			8270	1	1	*		4935	260		1255			F-1/DDH02-05
Chemex			8270	1	1	*		4017	336		1317			F-1/DDH02-05
Eco-tech			8273	1	1	*		2104	784		1671			F-1/DDH02-05
Chemex			8273	1	1	*		2847	675		1664			F-1/DDH02-05
Eco-tech			8279	1	1	*		3189	298		3522			F-1/DDH02-05
Chemex			8279	1	1	*		2658	260		3250			F-1/DDH02-05

Legend

Sample type

A - Original Sample

B - Duplicate Sample

C -Check Sample

D - Pulp Check Sample

E -Reject Check Sample

F - Lab recheck

Analysis Type

1 - ICP

2 - Fire Assay

2A - Metalic fire assay

3 - Metal Assay

Mineral Titles Tenure Search Results

Last updated April 30, 2002

17 Matches

Criteria	Owner Name	Tenure Type	Tenure Status
	BARKER MINERALS	Mineral	Good Standing

Tenure Number	Claim Name	Owner Number	Map Number	Work Recorded To	Status	Mining Division	Area	Tag Number
339981	ZAK 1	140410 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	18 un	223501
339982	ZAK 2	140410 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	18 un	223502
339983	ZAK 3	140410 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	18 un	223503
339984	ZAK 4	140410 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	18 un	223504
339985	ZAK 5	140410 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	16 un	223505
387808	JAKK 1	140410 100%	093B16E	2003.06.22	Good Standing 2003.06.22	03 CARIBOO	20 un	240446
387809	JAKK 2	140410 100%	093B16E	2003.06.22	Good Standing 2003.06.22	03 CARIBOO	20 un	240447
387810	JAKK 3	140410 100%	093B16E	2003.06.21	Good Standing 2003.06.21	03 CARIBOO	20 un	240448
391830	ROL #1	140410 100%	093A11W	2003.02.03	Good Standing 2003.02.03	03 CARIBOO	9 un	244180
391831	ROL #2	140410 100%	093A11W	2003.02.03	Good Standing 2003.02.03	03 CARIBOO	10 un	244181
391832	ROL #3	140410 100%	093A11W	2003.02.03	Good Standing 2003.02.03	03 CARIBOO	20 un	244182
392031	SC 1	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	12 un	244183
392032	SC 2	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	20 un	244184
392033	SC 3	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	12 un	244185
392034	SC 4	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	20 un	244186
392035	SC 5	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	20 un	244187
392036	SC 6	140410 100%	093A11W	2003.02.12	Good Standing 2003.02.12	03 CARIBOO	20 un	244188

Mineral Titles Tenure Search Results

Last updated April 30, 2002

400 Matches	Criteria	Owner Name	Tenure Type	Tenure Status
		DOYLE, LOUIS	Mineral	Good Standing

Tenure Number	Claim Name	Owner Number		Map Number	Work Recorded To	Status	Mining Division		Area		Tag Number
322616	UNLIKELY I	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1	un	655356M
322617	UNLIKELY II	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1	un	655357M
322720	ACE 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650242M
322721	ACE 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650243M
322722	ACE 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650244M
322723	ACE 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650245M
322724	ACE 5	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650246M
322725	ACE 6	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650247M
322726	ACE 7	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650248M
322727	ACE 8	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650249M
322728	ACE 9	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650250M
322729	ACE 10	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650301M
322730	ACE 11	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650302M
322731	ACE 12	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650303M
322732	ACE 13	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650304M
322733	ACE 14	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650305M
323065	ACE 15	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650251M
323066	ACE 16	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650252M
323067	ACE 17	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650253M
323068	ACE 18	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650254M
323069	ACE 19	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650255M
323070	ACE 20	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650256M
323071	ACE 21	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650257M
323072	ACE 22	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650258M
323073	ACE 23	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650259M
323074	ACE 24	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650260M

323075	ACE 25	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650261M
323076	ACE 26	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650262M
323077	ACE 27	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	647220M
323078	ACE 28	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650310M
323079	ACE 29	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650311M
323080	ACE 30	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	619897M
323081	ACE 31	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	619898M
323082	ACE 32	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	619899M
323083	ACE 33	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	619900M
323084	ACE 34	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	620207M
323085	ACE 35	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	620208M
323086	ACE 36	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	618899M
323087	ACE 37	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650355M
323088	ACE 38	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650338M
323089	ACE 39	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650339M
323090	ACE 40	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650340M
323091	ACE 41	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650341M
323092	ACE 42	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650342M
323093	ACE 43	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650343M
323094	ACE 44	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	650344M
331316	ACE 57	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1	un	628362M
331317	ACE 58	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628363M
331318	ACE 59	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628364M
331319	ACE 60	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628365M
331320	ACE 61	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628366M
331321	ACE 62	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628367M
331322	ACE 63	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628368M
331323	ACE 64	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628369M
331324	ACE 65	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628370M
331325	ACE 70	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628385M
331326	ACE 71	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628386M
331327	ACE 72	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628387M
331328	ACE 73	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1	un	628388M

331329	ACE 74	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	628389M
331330	ACE 75	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	628390M
331331	ACE 76	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	628391M
331332	ACE 77	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647601M
331333	ACE 78	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647602M
331334	ACE 79	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647603M
331335	ACE 82	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	647609M
331336	ACE 83	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	647610M
331337	ACE 84	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	647611M
331338	ACE 85	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	647612M
331501	ACE 86	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647613M
331502	ACE 87	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647614M
331503	ACE 88	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647615M
331504	ACE 89	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647616M
331505	ACE 90	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647617M
331506	ACE 91	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647618M
331507	ACE 92	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647619M
331508	ACE 93	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647620M
331509	ACE 94	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	647621M
331510	ACE 95	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628392M
331511	ACE 96	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628393M
331512	ACE 97	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628394M
331513	ACE 98	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628395M
331514	ACE 99	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628396M
331515	ACE 100	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628397M
331516	ACE 101	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628398M
331517	ACE 102	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628399M
331518	ACE 103	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628400M
331519	ACE 104	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628372M
331520	ACE 105	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628373M
331521	ACE 106	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628374M
331522	ACE 107	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628375M
331523	ACE 108	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628376M

331524	ACE 109	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628377M
331525	ACE 110	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628378M
331526	ACE 111	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628379M
331527	ACE 112	134602	100%	093A14E	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	628380M
332097	ACE 67	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647606M
332098	ACE 68	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647607M
332099	ACE 69	134602	100%	093A14E	2005.11.30	Good Standing 2005.11.30	03	CARIBOO	1 un	647608M
332104	LED 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647681M
332105	LED 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647682M
332106	LED 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647683M
332107	LED 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647684M
332108	LED 5	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647685M
332109	LED 6	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647686M
332110	LED 7	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647687M
332111	LED 8	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647688M
332112	LED 9	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647689M
332113	LED 10	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647690M
332114	LED 11	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647691M
332115	LED 12	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647692M
332116	LED 13	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647693M
332117	LED 14	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647694M
332118	LED 15	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647695M
332119	LED 16	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647696M
332120	LED 17	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647697M
332121	LED 18	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647698M
332122	LED 19	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647699M
332123	LED 20	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	647700M
332124	LED 21	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	625781M
332125	LED 22	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	625782M
332126	LED 23	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	625783M
332127	LED 24	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	625784M
334101	JIM	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15 un	200891
334102	KATHY	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15 un	200890

334103	COL 6	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	10 un	200888
334104	LD 1	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	6 un	200449
334105	LD 2	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	12 un	200450
334106	AARON	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	15 un	200892
334107	KIM	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	15 un	200889
334108	COL 5	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	10 un	200887
334284	E 1	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	625785M
334285	E 2	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	625786M
334286	E 3	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	625787M
334287	E 4	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	625788M
334288	E 5	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	625789M
335484	ROAR 1	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	6 un	213587
335485	ROAR 2	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	12 un	213588
335486	ROAR 3	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	9 un	213589
335487	ROAR 4	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	18 un	213575
335600	E 6	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	1 un	650351M
335601	PRINCE 1	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	15 un	213591
335602	PRINCE 2	134602 100%	093A14E	2006.11.30	Good Standing 2006.11.30	03 CARIBOO	20 un	213586
335603	ABRACAD 1	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	213594
335604	ABRACAD 2	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	9 un	213595
335605	KING 1	134602 100%	093A14E	2003.11.30	Good Standing 2003.11.30	03 CARIBOO	20 un	213592
335606	QUEEN 1	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	213593
336739	ACE WEST 1	134602 100%	093A14E	2006.11.30	Good Standing 2006.11.30	03 CARIBOO	20 un	210389
336740	ACE WEST 2	134602 100%	093A14E	2006.11.30	Good Standing 2006.11.30	03 CARIBOO	20 un	210489
338226	BILL 1	134602 100%	093A14E	2005.11.30	Good Standing 2005.11.30	03 CARIBOO	20 un	210861
338228	CHARLIE 1	134602 100%	093A14E	2004.11.30	Good Standing 2004.11.30	03 CARIBOO	1 un	647674M
338229	CHARLIE 2	134602 100%	093A14E	2004.11.30	Good Standing 2004.11.30	03 CARIBOO	1 un	647675M
338230	CHARLIE 3	134602 100%	093A14E	2004.11.30	Good Standing 2004.11.30	03 CARIBOO	1 un	647676M
338231	CHARLIE 4	134602 100%	093A14E	2004.11.30	Good Standing 2004.11.30	03 CARIBOO	1 un	647677M
338232	CHARLIE 5	134602 100%	093A14E	2004.11.30	Good Standing 2004.11.30	03 CARIBOO	1 un	647678M
343725	CHAY 1	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	210818
343726	CHAY 2	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	210819
343727	CHAY 3	134602 100%	093A14E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	210820

343728	CHAY 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210821
343735	RIVY 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210828
343736	RIVY 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210829
343737	TYS 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210830
343738	TYS 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210831
343745	TYS 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210832
343746	TYS 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210833
343751	NET 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10	un	210896
343752	NET 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210897
343753	NET 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	210898
343754	NET 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	210899
343755	BOO 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	210816
343756	BOO 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210817
343757	BLACK 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	210822
343758	BLACK 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210823
343759	GAR 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	5	un	210834
343761	GAR 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210835
343766	CATH 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210824
343767	CATH 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	210825
343768	CATH 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	210826
343769	CATH 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15	un	210827
343986	BROWN 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15	un	213601
343987	BROWN 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	213602
343988	BROWN 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213603
343989	BROWN 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213604
343990	SELL 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	6	un	213605
343991	SELL 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	4	un	213606
343992	SELL 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	8	un	213607
343993	SELL 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	213608
344004	BAD 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210846
344005	BAD 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210847
344006	BAD 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210848
344007	BADGER 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	6	un	210850

344008	BADGER 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	14	un	210864
344009	STEVEN 1	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213609
344010	STEVEN 2	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213610
344011	STEVEN 3	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10	un	213611
344012	STEVEN 4	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10	un	213612
344013	SON 1	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15	un	208212
344014	SON 2	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	208213
344015	GOO 1	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	210849
345690	AUBAR 1	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	8	un	213621
345691	AUBAR 2	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15	un	213622
345692	AUBAR 3	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213623
345693	AUBAR 4	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213624
345694	AUBAR 5	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213625
345695	AUBAR 6	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	213626
345696	AUBAR 7	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213627
345698	AUBAR 9	134602	100%	093A15W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	213629
345699	AUBAR 10	134602	100%	093A15W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213630
345702	AUBAR 13	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213633
345703	AUBAR 14	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	213634
346689	CHRIS 4	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	213656
346690	CHRIS 5	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	213657
346691	CHRIS 6	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213658
346692	CHRIS 7	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213659
346693	CHRIS 8	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213660
346694	CHRIS 9	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	213661
346695	CHRIS 10	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213662
346696	CHRIS 11	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	213663
347057	COMET 8	134602	100%	093A14E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	9	un	210865
347062	AMANDA 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213664
347063	AMANDA 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	213665
347064	AMANDA 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18	un	213666
347065	AMANDA 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213667
347066	AMANDA 5	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	213613

347067	AMANDA 6	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	213614
347068	AMANDA 7	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	213615
347069	AMANDA 8	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	4 un	210900
347222	GRAIN 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231131
347223	GRAIN 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16 un	231132
347224	GRAIN 3	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231133
347225	GRAIN 4	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231134
347226	GRAIN 5	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16 un	231135
347227	GRAIN 6	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16 un	231136
347483	B.B. 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231141
347484	B.B. 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231142
347485	B.B. 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231143
347486	B.B. 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231144
347589	B.B. 5	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231145
347590	B.B. 6	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231146
347591	B.B. 7	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10 un	231147
347592	B.B. 8	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231148
347593	B.B. 9	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10 un	231149
347595	B.B. 11	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231130
347964	JESS 1	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	18 un	231137
347965	JESS 2	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	18 un	231138
347966	JESS 3	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	18 un	231139
347967	JESS 4	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	18 un	231140
347968	LONG 1	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231171
348637	LONG 3	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208467
348639	LONG 5	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208469
348640	LONG 6	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208470
348641	GRACE 1	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208471
348642	GRACE 2	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208472
348643	GRACE 3	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	208473
348644	GRACE 4	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	5 un	208474
348645	GRACE 5	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	18 un	208475
348646	GRACE 6	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15 un	208476

348647	GRACE 7	134602	100%	093A11E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12 un	208462
351089	BIG GULP 1	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	12 un	208481
351090	BIG GULP 2	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	15 un	208482
351091	BIG GULP 3	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	12 un	208483
351092	BIG GULP 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15 un	208484
369406	FRANK	134602	100%	093A14W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	20 un	236768
369407	F1	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686507M
369408	F2	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686508M
369409	F3	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686509M
369410	F4	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686510M
369411	F5	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686511M
369412	F6	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686512M
369413	F7	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686513M
369414	F8	134602	100%	093A11W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	686514M
373146	FE1	134602	100%	093A14W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	692761M
373147	FE2	134602	100%	093A14W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	692762M
373148	FE3	134602	100%	093A14W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	692753M
373149	FE4	134602	100%	093A14W	2003.11.30	Good Standing 2003.11.30	03	CARIBOO	1 un	692754M
373150	FE5	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1 un	692755M
373333	FE	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	231935
375715	K1	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239915
375716	K2	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239916
375717	K3	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239917
375718	K4	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239918
375748	K5	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239925
375756	K6	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239926
375757	K9	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239929
375758	K10	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239930
375759	K11	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239921
375760	K12	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239922
375761	K13	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239923
375762	K7	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239927
375763	K8	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20 un	239928

376209	M 1	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	692760M
376210	M 2	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	691374M
376211	M 3	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	691373M
376212	MAUD 1	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	203376
376213	MAUD 2	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	203377
376214	MAUD 3	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	203378
376215	MAUD 4	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	203379
376216	MAUD 5	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	203380
376217	MAUD 6	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239919
377020	MAUD 10	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239951
377021	MAUD 11	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239952
377022	MAUD 15	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	15 un	239956
377023	MAUD 16	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	12 un	239957
377256	K18	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697519M
377257	K19	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697520M
377258	K20	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697521M
377259	MAUD 7	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239968
377260	MAUD 8	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239969
377261	MAUD 9	134602	100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239970
377431	FK 20	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	6 un	239920
377515	BEAR 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	237845
377516	BEAR 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	237846
377517	BEAR 3	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	237880
377688	VC 1	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238001
377720	VC 8	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	16 un	238078
377721	VC 9	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238079
377724	VC 6	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238006
377725	VC 7	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238007
377726	VC 2	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238002
377727	VC 3	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238003
377728	VC 5	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238005
377729	VC 11	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238081
377730	VC 12	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238082

377731	VC 13	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	238083
377732	VC 4	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	238004
377733	VC 10	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237924
377878	MAG A	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697573M
377879	MAG B	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697574M
377880	MAG C	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697575M
377881	MAG D	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697576M
377882	MAG E	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697577M
377883	MAG 1	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237896
377884	MAG 3	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237898
377885	MAG 5	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	10	un	237894
377899	MAG 2	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237897
377987	MADAM 1	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237849
377988	MADAM 2	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237850
377989	MADAM 3	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	237876
377990	MADAM 4	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	237877
377991	MADAM 5	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	237878
377992	MADAM 6	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	237879
377993	MADAM 7	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237847
377994	MADAM 8	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	237848
378113	P.G.	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	238072
378114	PG 1	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239981
378115	PG 2	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239982
378116	PG 11	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	238073
378117	PG 12	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	238074
378118	PG 13	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	238075
378119	PG 14	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	238076
378120	PG 4	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239984
378121	PG 5	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239985
378122	PG 6	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239986
378123	PG 7	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239987
378124	P.G 8	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	238066
378125	COTT 1	134602	100%	093A13W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	697522M

378126	COTT 2	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697523M
378127	COTT 3	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697528M
378128	COTT 4	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697529M
378129	COTT 5	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697530M
378130	COTT 6	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697531M
378131	COTT 7	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697532M
378132	COTT 8	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697533M
378133	COTT 9	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697534M
378134	COTT 10	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697535M
378135	COTT 11	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697536M
378136	COTT 12	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697537M
378137	COTT 13	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697538M
378138	COTT 14	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697539M
378143	COTT 16	134602 100%	093A12E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697541M
378144	COTT 17	134602 100%	093A12E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697567M
378145	PG 3	134602 100%	093A12E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239983
378146	P.G. 10	134602 100%	093A12E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	239980
378148	COTT 15	134602 100%	093A13W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697540M
378289	MAUD	134602 100%	093A12W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238084
378295	PG 16	134602 100%	093A13E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	4 un	237950
378371	MAG F	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697578M
378372	MAG G	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697579M
378373	MAG H	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697580M
378374	MAG I	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697581M
378375	MAG J	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697582M
378376	MAG K	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697583M
378377	MAG L	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697584M
378378	MAG M	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697585M
378379	MAG N	134602 100%	093B16E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	697586M
379267	K 15	134602 100%	093A12E	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	20 un	238085
379270	K 14	134602 100%	093A11W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	14 un	237937
379274	K 21	134602 100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	699911M
379275	K 23	134602 100%	093A14W	2002.11.30	Good Standing 2002.11.30	03 CARIBOO	1 un	699913M

379276	K 25	134602	100%	093A14W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	699915M
379556	PG 9	134602	100%	093A12E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	239979
379557	PG 15	134602	100%	093A13E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	16	un	237930
379810	CU2S 1	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	215159
379811	CU2S 2	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	215189
379812	CU2S 3	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	20	un	215160
379813	CU2S 4	134602	100%	093B16E	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	15	un	215190
382062	BIG GULP 5	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	9	un	203888
382063	BIG GULP 6	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	9	un	203889
382064	BIG GULP 7	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	12	un	210485
382172	BIG GULP 8	134602	100%	093A11W	2002.11.30	Good Standing 2002.11.30	03	CARIBOO	1	un	695800M