

LOG NO:	FEB 10	R.D.
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**ASSESSMENT REPORT
ON
GEOLOGICAL & GEOCHEMICAL WORK
ON THE FOLLOWING CLAIMS**

CROESUS 1 6129(5)
 CROESUS 2 6130(5)
 CROESUS 3 6131(5)
 CROESUS 4 6132(5)

NOTICE TO GROUP #3009712

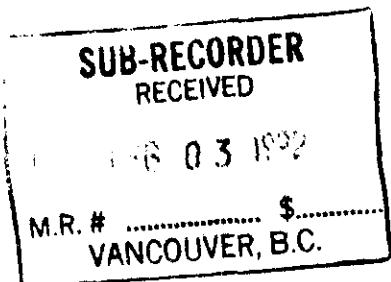
located

34 KM EAST OF
 STEWART, BRITISH COLUMBIA
 SKEENA MINING DIVISION

56 degrees 00 minutes latitude
 129 degrees 31 minutes longitude

N.T.S. 104A/4E, 104A/3W, 103P/13E & 103P/14W

PROJECT PERIOD: July 1 to Oct. 7, 1991



ON BEHALF OF
 TEUTON RESOURCES CORP.
 VANCOUVER, B.C.

REPORT BY

D. Cremonese, P. Eng.

509-675 W. Hastings
 Vancouver, B.C.

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

Date: January 31, 1992

22,103

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1. INTRODUCTION

A. Property, Location, Access and Physiography

The property is located about 34 km east of Stewart, British Columbia. Nearest paved road is the Bear River Highway about 13 km to the north. Access is presently limited to helicopter, either from the base at Stewart (Vancouver Island Helicopters) or from the Ellsworth Logging Camp on Highway 13 (Highland Helicopters). The latter is generally more suitable both because of its greater proximity to the property and also because of the absence of high, intervening mountains. There is a possibility that logging roads running west across the Nass River from Highway 13 may one day provide the closest approach to the property.

The Croesus claims are centered roughly at the foot of Del Norte Glacier, which flows east out of the Cambria Icefield and gives rise to Del Norte Creek. Elevations vary from approximately 1050 meters on the creek bed at the eastern edge of the property to more than 2000 meters near ridge tops. Vegetation in the area changes from a mantle of mountain hemlock and balsam at low-lying elevations to shrubs, mountain grasses and heather at higher elevations. Slopes range from moderate to steep to precipitous.

Climate is relatively severe, particularly at higher elevations. Because the property lies on the eastern edge of the Cambria Icefield, precipitation is not as pronounced as in the immediate Stewart area.

B. Status of Property

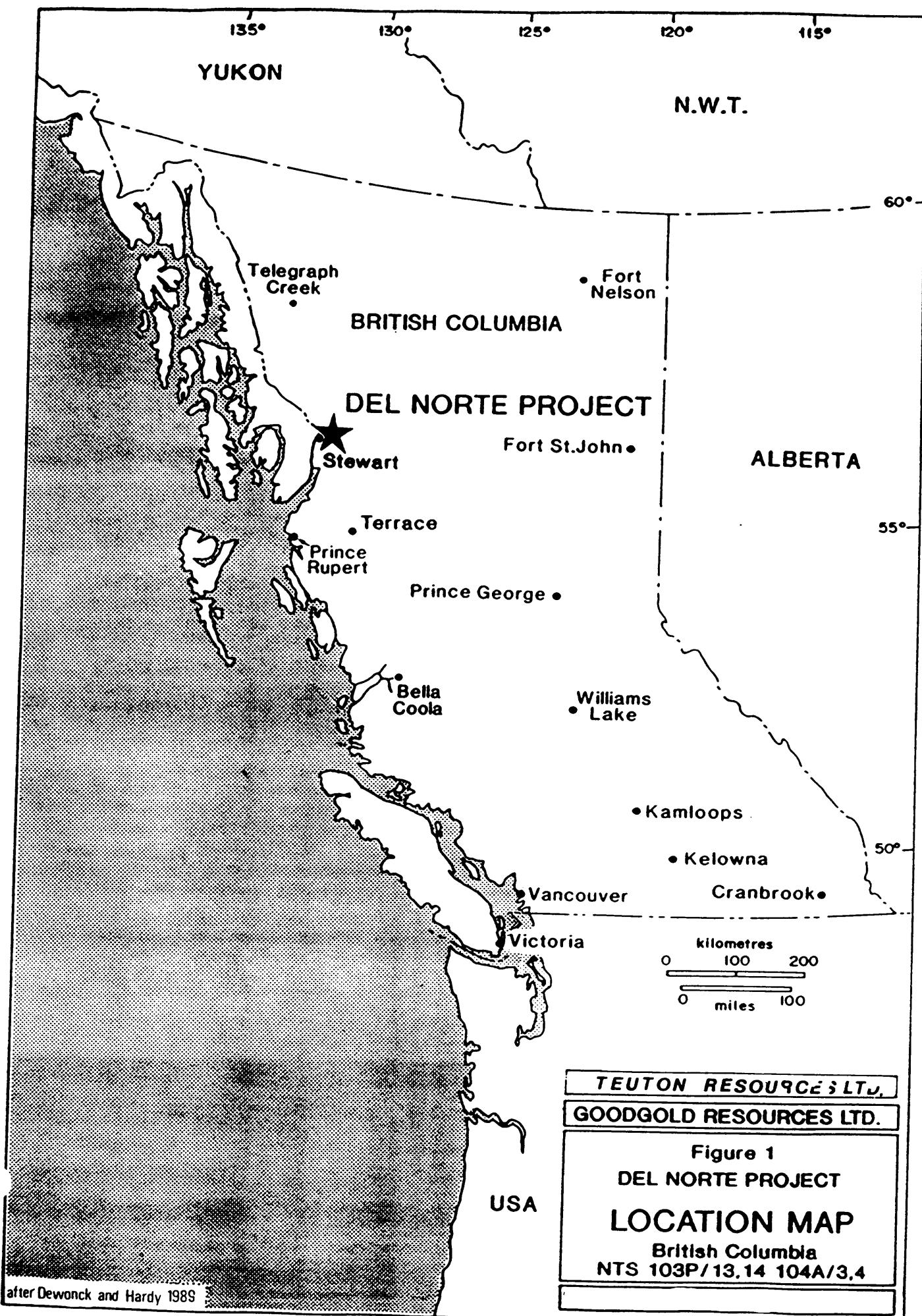
Relevant claim information is summarized below:

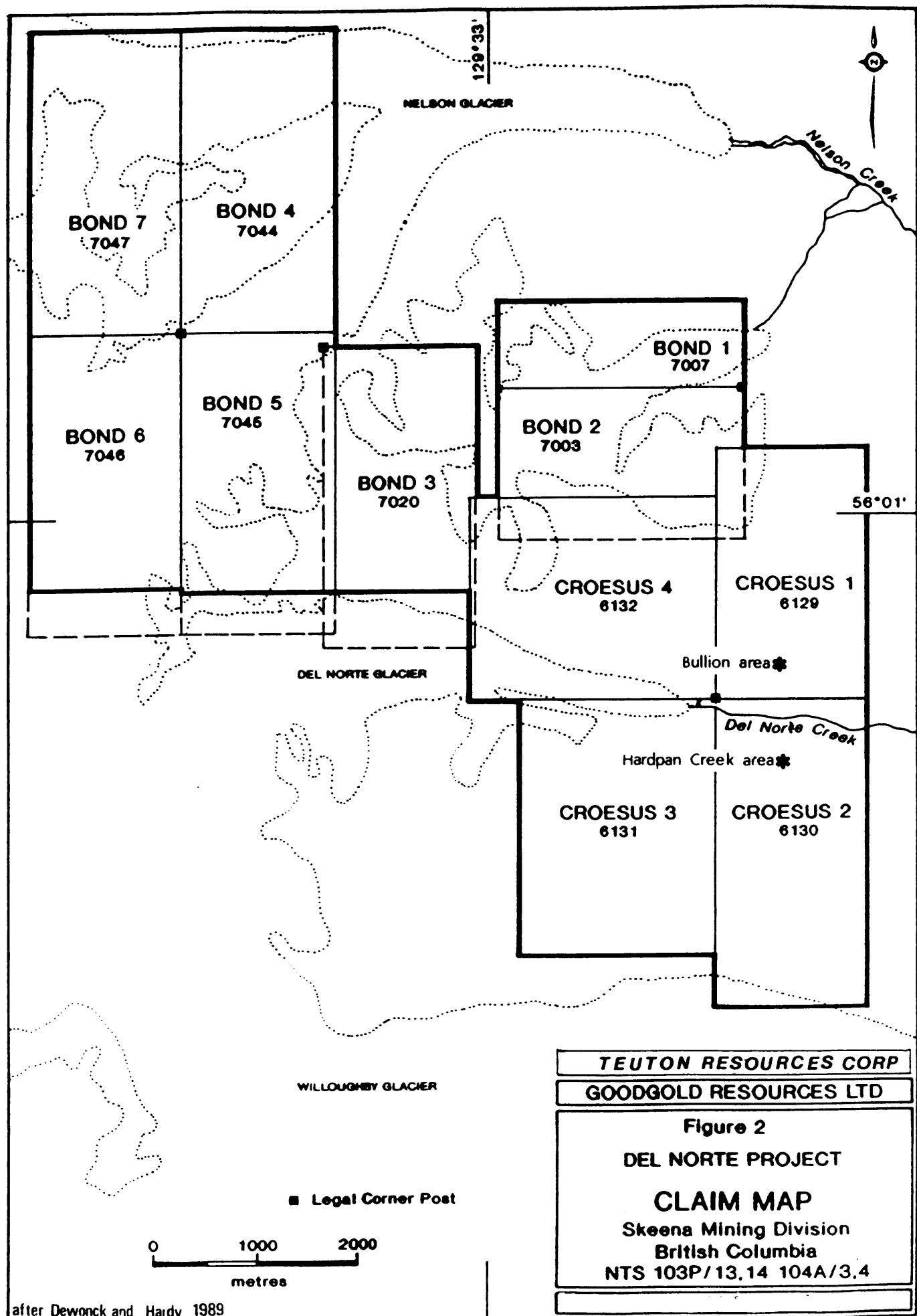
Name	Record No.	No. of Units	Record Date
Croesus 1	6129(5)	15	May 4, 1987
Croesus 2	6130(5)	18	May 4, 1987
Croesus 3	6131(5)	20	May 4, 1987
Croesus 4	6132(5)	20	May 4, 1987

Claim locations are shown on Fig. 2 after government N.T.S. maps. The claims are owned by Teuton Resources Corp. of Vancouver, British Columbia. During the first part of the 1991 assessment work program the claims were under option to Goodgold Resources Ltd. of Vancouver; Goodgold relinquished its option in August of 1991. Further work was financed by Teuton Resources Corp.

C. History

Records indicate that the property was originally staked as the "Bullion" claim, sometime prior to 1913. This early work was





undoubtedly a follow-up to the small-scale placer gold operations reported to have taken place on Nelson, Del Norte and Willoughby Creeks.

Between this first staking and 1922, when the property was restaked as the Delnorte Group by Green and Ficklin of Hyder, Alaska, a small adit was driven to test a zone of quartz veining paralleling the contact between Bowser sediments and Hazelton volcaniclastics (on the north side of Del Norte Creek, Croesus 1 claim: cf. Bullion grid, *supra*). In 1939, Owen McFadden of Stewart, backed by a syndicate, explored the ground by a series of fifteen open-cuts and some small popholes. At this time the property was known as the "Meziadin Group". In the same year, the property was visited by Dr. Mandy of the B.C. Department of Mines; Mandy examined and sampled several of the showings. Samples results indicated erratic low-grade gold mineralization associated with copper and occasional zinc values (Ref. 7, 1939). According to extant records, most of the sampling was from the north side of Del Norte Creek.

Exploration carried out during this period was severely restricted by difficult access. The trail leading into the Del Norte Creek drainage from the end of the Bear River road was over 75 km long and entailed two difficult mountain crossings.

In the 1960's the area was explored again by companies searching for porphyry copper deposits. This, and subsequent work, was supported by helicopter. In the late 1970's and early 1980's, renewed exploration efforts concentrated on precious metals. Apparently, this work did not uncover anything of importance in the Del Norte Creek area (Ref. 6).

In 1987 Teuton Resources Corp. acquired the Croesus claims and carried out a program of rock and silt sampling (Ref. 9). Silt samples taken from the creek draining the Bullion showing returned moderate to highly anomalous values in gold, silver, copper, lead, and zinc. The best rock grab sample assayed 19,300 ppb Au and came from a quartz sulfide lens in a prominent gossan on the southern side of Del Norte Creek (Hardpan Creek area).

In 1988 Teuton followed up on these results with a limited program of geological mapping, prospecting, rock sampling and soil sampling in the Bullion and Hardpan Creek areas (Ref. 10). Two zones, one featuring lead-zinc mineralization, the other copper-gold, were discovered in the Hardpan Creek drainage. Several grab samples taken peripheral to these zones returned anomalous values in gold, silver, copper, lead and zinc.

On the strength of the 1988 work, and collaterally because of the enthusiasm generated by the major Eskay Creek discoveries, Teuton was able to option the property to Goodgold Resources Ltd. in 1989. During 1989, Goodgold contracted Aerodat (Ref. 11) to

carry out an airborne EM and Magnetometer survey over the property. Results outlined a magnetically higher central area (corresponding to volcanic rocks, and/or intrusives) flanked on the northwest and east by a lower slowly varying magnetic field (corresponding to sedimentary rocks). Goodgold also completed a small surface program concentrating on the Bullion area, with mixed results (Ref. 12).

In 1990, Goodgold mounted a major \$500,000+ program focussing mostly on the Hardpan Creek portion of the property and consisting of a preliminary phase of grid construction, mapping/prospecting, blasting/trenching, soil geochemical sampling, and geophysical surveying, followed by a second phase of diamond drilling entailing 12 holes (total 1,119m). Results of this work were compiled in a lengthy report by Bishop and Gal (Ref. 13, on file with BCEMPR).

References

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E. Summary of Work Done.

The bulk of the 1991 geological and geochemical program on the property was undertaken by contractor Nicholson & Associates under the supervision of the author. The project spanned the period from July 1 to October 7, 1991.

Crew, camp, equipment and supplies were mobilized by helicopter from the Ellsworth Logging Camp on Highway 13. The first area to be investigated was on the north side of Del Norte Creek, the "Bullion Grid" area (cf. Figs. 5-14), opposite the Hardpan Creek zones explored in 1990. A grid was established with maximum dimensions of 1,200 m (N-S) and 800 m (E-W). A soil geochem program over the Bullion grid consisted of 431 samples; rock sampling consisted of 149 samples. Geological mapping was also undertaken in this area.

A few weeks into the program, one of the geologists discovered a possible porphyry gold-copper zone or zones about 1km west of the Bullion Grid. Subsequently some reconnaissance geochem soil lines were emplaced in this area, prospecting and silt samples taken, and trench and diamond saw rock samples taken. All told 148 soil samples, 71 rock samples and 22 silt samples were collected from the "Porphyry" area (cf. Figs. 15-16).

Follow-up trenching and sampling were also carried out on the Hardpan Creek side on the Humdinger zone (97 rock samples) and the Grizzly zone (32 samples). Results are presented in Fig. 17.

During the latter part of the program, a geophysical crew was also brought in to carry out IP and Mag surveys over the Porphyry Zone and EM/Mag surveys over the Bullion Grid area. Because this work was completed with scattered coverage only, the data has not been entered into this report. However, on the Bullion Grid Compilation map (Fig. 14) the outline of a local magnetic anomaly has been included because it coincides well with certain geochemical soil anomalies. [Note: None of the costs or expenses associated with any of the geophysical surveys have been incorporated within the work cost statement.]

All samples taken during the 1991 assessment work program were analysed for gold by standard AA techniques, as well as for 30 elements by I.C.P. (Inductively Coupled Argon Plasma) at the Eco-Tech facility in Kamloops, B.C. Samples containing high values in silver or base metals were subjected to wet chemical assays and high golds were additionally tested by the metallics method to determine whether coarse gold was present.

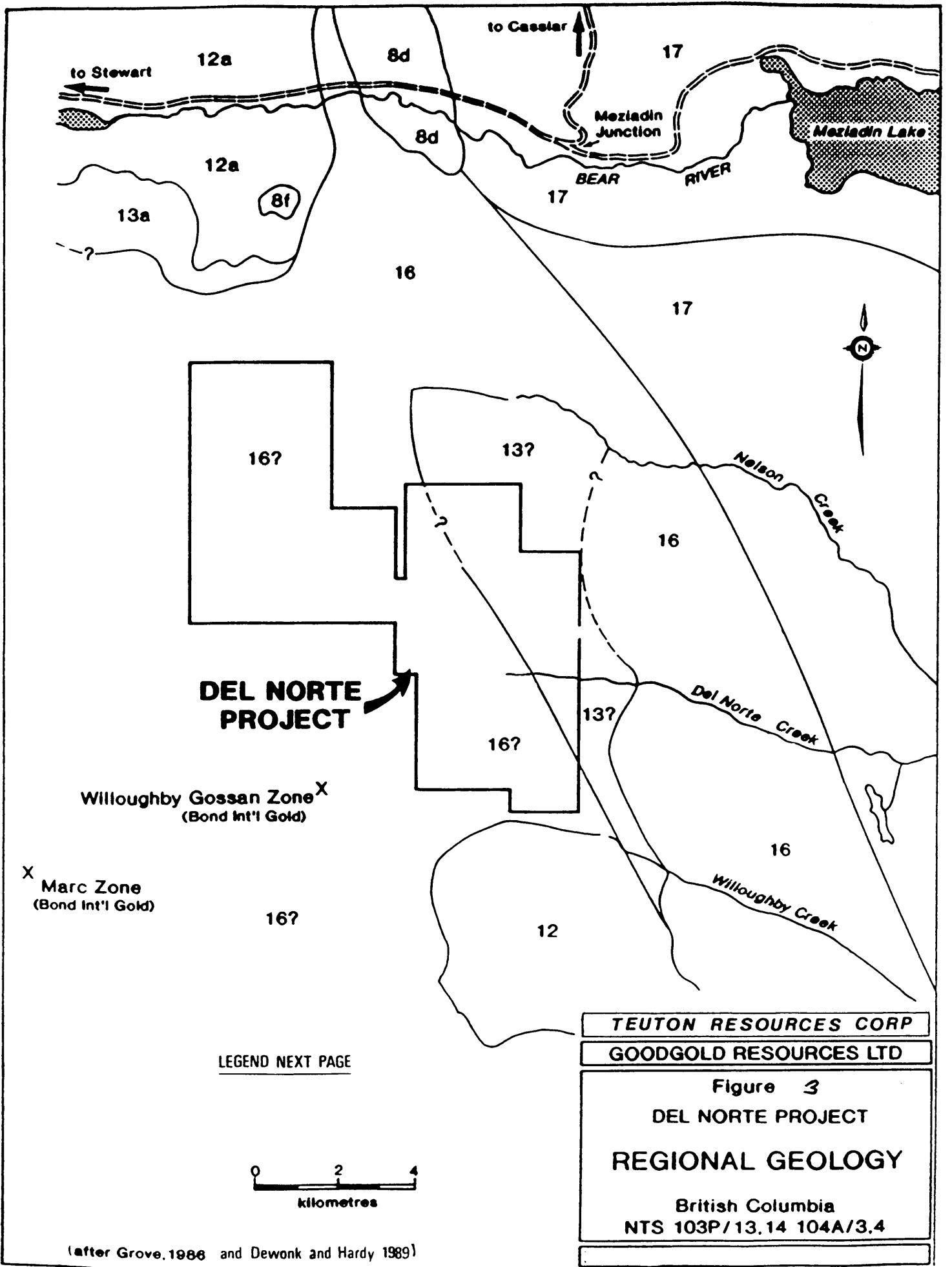
2. TECHNICAL DATA AND INTERPRETATION

A. Regional Geology

The property lies along the eastern edge of a broad, north-northwest trending belt of Triassic and Jurassic volcanic and sedimentary rocks termed by Grove (1971) as the "Stewart Complex". This belt is bounded to the west by the Coast Crystalline Belt (mainly granodiorites) and to the east by a thick series of sedimentary rocks known as the Bowser Assemblage (Middle Jurassic to Upper Jurassic age).

A major contact between sedimentary rocks of the Bowser Group and volcanics of the lower Jurassic Hazelton Group passes north-south between Strohn Creek and the White River. In between these two watercourses are three west-east flowing streams originating in the Cambria icefield and all carrying placer gold. These streams, from north to south, are Nelson Creek, Del Norte (also known as "Porter") Creek and Willoughby Creek. The source of the placer gold has intrigued Stewart area prospectors for many years. Although Bond Gold's Marc Zone gold discovery at Red Mountain, about 12 km west of the property, has focussed some recent attention on the area, it has till now received little attention from government geologists and the best studies are in private reports. The author was able to locate a good description of regional geology in this area from such a report--a lengthy excerpt from Downing (1983) follows:

"Tectonically, the Bowser-Hazelton contact appears to be a thrust zone with Bowser sediment "slices" occurring within and overlying the Hazelton volcanics to the west. No Hazelton rocks were noted overlying the Bowser sediments to the east. The



LEGEND for Figure 4

SEDIMENTARY AND VOLCANIC ROCKS (after Grove, 1986)

QUATERNARY

RECENT

20 UNCONSOLIDATED DEPOSITS: RIVER FLOODPLAIN, ESTUARINE, RIVER CHANNEL AND TERRACES, ALLUVIAL FANS, DELTAS AND BEACHES, OUTWASH, GLACIAL LAKE SEDIMENTS, TILL, PEAT, LANDSLIDES, VOLCANIC ASH, HOTSPRING DEPOSITS

19 BASALT FLOWS (I), CINDER, ASH (I)

PLEISTOCENE AND RECENT

BASALT FLOWS

JURASSIC

HAZELTON GROUP

UPPER JURASSIC

NASS FORMATION

17 SILTSTONE, GREYWACKE, SANDSTONE, SOME CALCARENITE, ARGILLITE, CONGLOMERATE, MINOR LIMESTONE, MINOR COAL (INCLUDING EQUIVALENT SHALE, PHYLLITE, AND SCHIST)

MIDDLE JURASSIC

SALMON RIVER FORMATION

16 SILTSTONE, GREYWACKE, SANDSTONE, SOME CALCARENITE, MINOR LIMESTONE, ARGILLITE, CONGLOMERATE, LITTORAL DEPOSITS

15 RHYOLITE, RHYOLITE BRECCIA; CRYSTAL AND LITHIC TUFF

BETTY CREEK FORMATION

14 PILLOW LAVA, BROKEN PILLOW BRECCIA (I); ANDESITIC AND BASALTIC FLOWS (I)

13 GREEN, RED, PURPLE, AND BLACK VOLCANIC BRECCIA, CONGLOMERATE, SANDSTONE, AND SILTSTONE (I); CRYSTAL AND LITHIC TUFF (I); SILTSTONE (I); MINOR CHERT AND LIMESTONE (INCLUDES SOME LAVA (+14)) (I)

LOWER JURASSIC

UNUK RIVER FORMATION

12 GREEN, RED, AND PURPLE VOLCANIC BRECCIA, CONGLOMERATE, SANDSTONE, AND SILTSTONE (I); CRYSTAL AND LITHIC TUFF (I); SANDSTONE (I); CONGLOMERATE (I); LIMESTONE (I); CHERT (I); MINOR COAL (I)

11 PILLOW LAVA (I); VOLCANIC FLOWS (I)

TRIASSIC

UPPER TRIASSIC

TAKLA GROUP (I)

10 SILTSTONE, SANDSTONE, CONGLOMERATE (I); VOLCANIC SILTSTONE, SANDSTONE, CONGLOMERATE (I); AND SOME BRECCIA (I); CRYSTAL AND LITHIC TUFF (I); LIMESTONE (I)

PLUTONIC ROCKS

OLIGOCENE AND YOUNGER

9 DYKES AND SILLS (SWARMS), DIORITE (I); QUARTZ DIORITE (I); GRANODIORITE (I); BASALT (I)

EOCENE (STOCKS, ETC.) AND OLDER

8 QUARTZ DIORITE (I); GRANODIORITE (I); MONZONITE (I); QUARTZ MONZONITE (I); AUGITE DIORITE (I); FELDSPAR PORPHYRY (I)

7 COAST PLUTONIC COMPLEX: GRANODIORITE (I); QUARTZ DIORITE (I); QUARTZ MONZONITE, SOME GRANITE (I); MIGMATITE - AGMATIC (I)

CENOZOIC

MESOZOIC

CENOZOIC

Bowser sediments include shale, silt-mudstone, wacke and conglomerate while andesitic to rhyolitic tuffs and flows, limestone and argillite make up the Hazelton assemblage. The predominant dip direction of bedding in the Bowser sediments is northeasterly. Along the west fork to Surprise Creek, the Hazelton-Bowser contact is well preserved--tuffs and coarse tuff breccia overlain by a basal conglomerate grading to wacke-silt-mudstone-shale.

Several medium to coarse-grained porphyritic (potash feldspar) quartz monzonite and biotite granodiorite stocks occur along the contact zone. Other intrusives include augite to hornblende plagioclase porphyries of possible volcanic origin and northwest trending lamprophyre and hornblende porphyry dykes which in places form a dyke swarm, all of which occur predominantly south of the Stewart highway (Nelson-Porter-Willoughby Creeks area). [Note: Downing uses "Porter" to describe Del Norte Creek--this is an alternative name].

Metamorphism is predominantly of the greenschist facies on a regional scale. Andalusite occurs in the argillites on the west fork to Surprise Creek. Biotite hornfels zones are associated with a majority of the quartz monzonite-granodiorite stocks.

The east-west flowing Strohn and Bear Creeks (Stewart highway section) occur along a major tectonic break which transects the northerly trending structural fabric in the Stewart area. The sense and amount of displacement along this break (strike slip fault?) is unknown. Displacement along the Bowser-Hazelton contact in the Willoughby-Bowser Lake area is unknown, however, offset along this contact on the Long Lake fault north of Stewart indicates approximately 1500 feet (Grove, 1971). A dominant pyritic shear zone up to ten meters across occurs near the Hazelton-Bowser contact from Willoughby to Porter Creeks."

Property location relative to regional geology is shown on Fig. 3.

B. Property Geology

The local geology of the property area was sketched by Dr. Mandy, B.C. Department of Mines in 1939 (Ref. 7). Mandy shows the major volcaniclastic-sediment (Hazelton-Bowser) contact running roughly north-south, about 1,000 m or so east of the Legal Post for the Croesus claims. The volcaniclastics are described as a sequence of andesitic breccia (some lava), andesite, andesite tuff and lava locally pyritized and silicified, carbonate tuff locally pyritized and transitional tuffs/argillites.

The Bullion Grid area was mapped in 1991 by geologist Brian Game at a scale of 1:2,000. Results are presented in this report in Fig. 5. (Location relative to claim boundaries may be ascertained by reference to Fig. 4--Index Map).

The mapping shows a roughly NNW trending strike to two major units, the Betty Creek Formation (on the west) and the overlying Salmon River Formation (on the east). The Betty Creek Formation has been subdivided into five sub-units: 3a--intermediate plagioclase porphyry flows (andesite); 3b--intermediate ash lapilli and plagioclase crystal tuffs; 3c--agglomerate; 3d--argillite; and, 3e--strongly phylllic-argillic altered volcanics. Similarly, the Salmon River Formation has been refined into four sub-units: 2a--argillite, laminated mudstone; 2b--cherty argillite; 2c--siltstone; and, 2d--chert pebble conglomerate. A third unit, the Ashman Formation, consisting of argillite and intraformational conglomerates outcrops in the southeast corner of the Bullion Grid. Several plagioclase hornblende porphyry dykes have also been mapped in this locality. Large portions of the grid area were not mapped because of extreme topography or talus cover.

Prospecting 1 km west of the Bullion Grid area disclosed a zone on the north side of the Del Norte Glacier marked by a series of blood-red discolored, resistive knobs jutting out of the glacial hardpan. Within this zone, a sub-area area was discovered and subsequently named the "Crackle Zone" because it featured a network of quartz stringers/veins (approx. 6 per 3m section), varying from 1 to 15 cm in width, within a silicified crystal tuff (Betty Creek Formation). The stringers/veins were observed to contain medium to coarse-grained inclusions of chalcopyrite, pyrite and to a lesser extent massive coarse-grained magnetite plus or minus arsenopyrite. Dip is generally 40-50 degrees to the west with a north-south strike. Observed outcrop of the Crackle Zone is about 50 by 100 m, possible extensions obscured by glacial hardpan, overburden and snow/ice. The Crackle Zone has been detailed at a scale of 1:200 in an inset map of Figs. 15-16. A large area surrounding the Crackle Zone is marked by pervasive propylitic and argillic alteration.

C. Geochemistry - Bullion Grid Area

a. Introduction

The Bullion Grid was constructed from an offset 1,200 m long, S-N baseline with cross-lines every 50m. Length of cross-lines varied according to topographical conditions, which were severe at many localities. Maximum E-W range of the cross-lines was 800m. Sample stations were every 25 m along the cross-lines.

The field crew collected 431 soil and 149 rock geochem samples from the Bullion Grid area. Grid location was fixed by reference to a base map prepared from an orthophoto.

b. Treatment of Data

Rock sample results have been presented in this report on Fig. 6 (Au, Ag & As Values) and Fig. 7 (Cu, Pb & Zn Values). Both

figures are at a scale of 1:2000; location relative to claim lines is indexed on Fig. 4. The majority of rock samples taken consisted of 1.0 m chip samples of outcrop. Continuous interval sampling at any one location has been duly noted in Figs. 6 and 7. Geochem sample values which were high enough to warrant further analysis by assay methods have been indicated in boldface.

Soil sample results are presented on Figs. 8 to 13, inclusive. These show, respectively, values for Au, Ag, As, Cu, Pb and Zn. Geochemical contours have been plotted using the Probplot program. Contour levels were chosen empirically, on a trial-and-error basis, the better to distinguish anomalous areas from background. This has been found to be a better method than rigorously applying standard statistical methods to the data set to define anomalous and non-anomalous categories.

The soil geochem maps were prepared to the same scale as the rock geochem maps and cover the identical area.

c. Discussion

The following observations were made after a review of the rock and soil geochem data from the Bullion Grid area:

- a. A strong Au-Pb-Zn-Cu (Ag) soil geochem anomaly is apparent in the western part of the grid between lines 4N and 6N, coincident with a N-S trending Mag anomaly. Similar geochem anomalies SSE in the lower part of the grid may represent a continuation of the source or downhill contamination. The intervening area between lines 2N and 4N was not sampled due to steepness of terrain.

Several of the rock samples taken in this area contained anomalous lead, zinc, gold and silver values, but nothing of exceptional grade. Whether the strong Pb and Zn soil geochem values (1000 ppm+) indicated in the core of the anomaly are explained by the outcrops already sampled, or are due to an undiscovered source, is not known. 1990 geochem surveys on the Hardpan Creek side generated similar Pb and Zn soil magnitudes but these were found to be associated with massive galena and sphalerite mineralization (cf. Twilight and Grizzly showings).

- b. A quartz sulfide vein, dubbed the "NMG Vein", was discovered in the NE quadrant of the grid, outcropping intermittently over a distance of 225 m. Like the LG vein discovered in 1990 to the north (on the Nelson Creek side), it lies along the volcanic-sedimentary contact. Best assay came from a 1m chip sample across the NMG vein at its southernmost exposure: 0.31 oz/ton gold and 16.67 oz/ton silver (#DM-MR-82646). The vein is tentatively associated with a sharp, flanking silver soil

anomaly between Lines 10.5N and 7.5N; a coincident gold anomaly is also apparent over a lesser interval.

The silver-gold anomaly located at the SE corner of the grid may reflect a southern continuation of the vein or another structure. Sample #82512 from this area ran 0.23 opt gold and 5.29 opt silver.

- c. Moderate to high-grade values were obtained from small veins exposed along the eastern bank of the central creek: see #'s 82502, 82503 and 82415. A select grab from the latter ran 0.75 opt Au, 52.21 opt Ag.

D. Geochemistry--Porphyry Zones

a. Introduction:

Discovery of a porphyry-type geological setting about 1 km west of the Bullion Grid area led to an expansion of the 1991 program. The first sub-area to be investigated was the so-called "Crackle Zone", outcropping in moraine just north of the toe of the Del Norte Glacier. Rock geochem reconnaissance samples were taken from this zone as well as outlying areas featuring similar mineralization, followed by diamond saw and trench sampling. Three soil contour lines were emplaced to test the area above the Crackle Zone.

An investigation of the large area south of Del Norte Glacier, just across the ice from the Crackle Zone, showed pervasive propylitic and argillic alteration. This area was tested by three long soil contour lines, silt sampling of two streams and minor rock geochem reconnaissance sampling.

Altogether 148 soil geochem samples, 71 rock samples and 22 silt samples were taken from the Porphyry zones. Grid location was fixed by reference to a base map prepared from an orthophoto.

b. Treatment of Data

Rock, soil, and silt geochem data is presented in this report in Fig. 15 (Sample Location Map) and Fig. 16 (Au, As & Cu Values). Both figures are at a scale of 1:5000. An inset map details the Crackle Zone; the majority of rock samples taken from diamond saw cuts consisted of 1.0 m interval samples of outcrop. Rock sample values which were high enough to warrant further analysis by assay methods are indicated in boldface.

The soil geochem data was not contoured because sampling was confined to irregular, often wide-spaced, topographical contour lines.

c. Discussion

Because the Crackle Zone appeared to be visually promising, most of the follow-up work was concentrated on this area (cf. Detail Map, Figs. 15-16). Unfortunately, this work was completed before assay values were in from the original reconnaissance samples: these unexpectedly showed low values in gold and low to fair values in copper.

As it turned out, follow-up detail work would have been more fruitful in another area of quartz stockworking located a few hundred meters to the southeast of the Crackle Zone. All five recon samples taken in this area returned gold values, ranging from 0.058 to 0.651 opt Au (the latter sample is #82532). At the close of the field season, some minor trenching was carried out in this area which confirmed the anomalous values in gold and arsenic obtained earlier. Samples ERK 277-285 from these trenches returned anomalous to ore grade values in gold as well as highly anomalous levels of arsenic.

Three recon geochem contour lines taken above the Crackle Zone returned numerous samples with elevated to anomalous levels in copper and arsenic, and to a lesser extent, gold. There is no definite trend apparent to the higher values, possibly because of the low density of coverage. Highest copper value of 1075 ppm came from the northwesternmost soil sample.

It is also noteworthy that three rock samples taken from the eastern limits of the soil geochem lines in this area returned anomalous values in tungsten (cf. especially #82534 registering 780 ppm W).

To the south of Del Norte Glacier, four silt samples (KM10-13: ETK 91-567) taken from a stream returned anomalous values in gold (45 to 195 ppb) and copper (147-177 ppm). Further south yet, on the Willoughby Creek slope, silt sampling returned similar anomalous values (KM 1-9 and MSS 28-36: ETK 91-567). The western 600 to 700 m of three contour soil lines in this area registered copper values averaging around 170 ppm, indicating a large area of elevated copper content. A few float samples taken further south, downhill, returned significant gold values (see, for instance, #81809: 0.081 oz/ton gold, 1% Cu.).

Brian Game, senior geologist during the Goodgold phase of the 1991 work, was of the opinion that the data pointed to a porphyry gold-copper system. Points advanced by Game in support of the hypothesis are: the quartz carbonate veining in the Crackle Zone is a typical manifestation of the top end of a porphyry system; propylitic and argillic alteration is pervasive over an extensive, surrounding area; the presence of widespread elevated copper geochemistry (accompanied in places by elevated gold values); the presence of fine-grained disseminated pyrite throughout the host

volcanics; the presence of coarse-grained magnetite; and, the presence of anomalous gold values peripheral to the central core.

E. Geochemistry--Humdinger and Grizzly Zones

a. Introduction

Further trenching was undertaken in both the Humdinger and Grizzly Zones, located on the Hardpan Creek side (site of almost all of the 1990 \$500,000+ work program). In all, 97 samples were taken from the Humdinger and 32 from the Grizzly: almost all of these represented 1m chip samples taken along trenches.

b. Treatment of Data

Results of 1991 trenching in both the Humdinger and Grizzly Zones are presented in Fig. 17. Each zone is depicted in a detail map at 1:100 scale, incorporated in a base map at a scale of 1:5000 which also shows claim boundaries.

For reference, the Humdinger Zone detail map also shows trenches emplaced in 1990 and associated assay values from interval sampling (these are not accompanied by any sample #'s so as to distinguish them from the 1991 assay values which are). Trench 90-3 was put in in 1990 but not sampled till 1991.

The Humdinger Zone detail map presents values for Au, Ag, Cu, Pb & Zn, the Grizzly Zone detail map for Au, Ag, Pb & Zn only (Cu is not anomalous in the Grizzly Zone).

c. Discussion

It was hoped that further trenching in the Humdinger Zone would discover mineralization similar to the interval obtained the previous year in Trench 90-2: 6 m of 0.388 opt gold. An attempt was made to try and extend the zone to the north through heavy and well-rooted overburden. The first of these extending trenches, #91-6, picked up the zone but it was not found in 91-7 further downhill to the north. Gold values were also obtained in what may be a parallel zone east of a fault located about 10 m east of the Humdinger zone (cf. 0.122 over 3 m--NE end of Trench 90-3: this trench was excavated in 1990 but not sampled till 1991). Most of the trenches were in highly oxidized material and efforts to go down far enough to fresh rock were largely unsuccessful. Sample #ERK 286 which returned 0.264 opt gold and 6.40 opt silver is a 1 m chip centred on 1990 recon sample DNSDR 039...it was not plotted on Fig. 17 through oversight.

Trenching on the Grizzly zone was carried out to better define attitude and extent of the massive Zn-Pb mineralization observed therein in 1990. Trenches were put in at right angles to the 1990

trenches, exposing mineralized "haloes" roughly at 45 degree orientation to the 1991 trenches. The best interval was from the northernmost trench, over 4m a weighted average of 0.062 opt Au, 3.92% Pb and 9.17% Zn. Assays from the more southerly trenches did not return similar values.

F. Field Procedure and Laboratory Technique

Rock samples were taken in the field with a prospector's pick and collected in a standard plastic sample bag. Grab samples were taken to ascertain character of mineralization at any specific locality. These samples consisted generally of three to ten representative pieces with total sample weight ranging between 0.5 to 2.0 kg. Chip samples were taken across the strike of mineralized structures and generally weighed about 1.0 to 2.0 kg. Interval samples from trenches were carefully taken to ensure a balanced weighting of sub-samples along the interval length.

Soil samples were taken in the field by digging with a mattock to the "C" soil horizon (poorly developed for the most part), with samples running approximately 300 to 500 grams of material. This was then placed into a standard Kraft Bag. The bags were then marked and allowed to dry before shipping.

Silt samples were taken from the active portions of the stream channels. Samples were carefully placed in standard Kraft Bags and allowed to dry before shipping.

All samples were analyzed at the Eco-Tech facility in Kamloops, B.C. Rock samples were first crushed to minus 10 mesh using jaw and cone crushers. Then 250 grams of the minus 10 mesh material was pulverized to minus 140 mesh using a ring pulverizer. For the gold analysis a 10.0 gram portion of the minus 140 mesh material was used. After concentrating the gold through standard fire assay methods, the resulting bead was then dissolved in aqua regia for 2 hrs at 95 deg. C. The resulting solution was then analysed by atomic absorption. The analytical results were then compared to prepared standards for the determination of the absolute amounts. For the determination of the remaining trace and major elements Inductively Coupled Argon Plasma (ICP) was used. In this procedure a 1.00 gram portion of the minus 140 mesh material is digested with aqua regia for 2 hours at 95 deg. C and made up to a volume of 20 mls prior to the actual analysis in the plasma. Again the absolute amounts were determined by comparing the analytical results to those of prepared standards.

Specific samples were subjected to further analysis where values obtained exceeded certain threshold levels. High golds were fire-assayed using conventional methods followed by parting and weighing of beads. Metallics assays were used in certain cases to test for the presence of coarse golds. Wet chemistry methods and

AA were used for follow-up analysis of base metals and silver (where values were too high for quantitative measurement by ICP).

Analysis of the soil and silt samples at the laboratory followed the same procedure as for the rocks, with the exception that sample preparation techniques were different (standard soil and silt sample preparation methods were used).

G. Conclusions

The 1991 work program on the Croesus claims outlined a number of areas deserving follow-up exploration. In the Bullion Grid area, a strong multi-element geochem anomaly in the southwest corner may signal a buried mineralized structure. Further prospecting and geophysical surveys are warranted to locate the source of this anomaly. In the northeast corner of the grid, the NMG quartz-sulfide vein has been shown to carry appreciable values in gold and silver. Continuity of this vein structure is as yet indeterminate and warrants further investigation. There is a possibility that the LG vein discovered in 1990 well to the north and in a similar environment may be a related structure. A sharp, flanking gold and silver soil geochemical anomaly may or may not be related to the NMG vein. More ground work is needed to test this hypothesis. The soil grid should be extended to the south of the vein to determine whether the anomalies continue. Samplers will probably need ropes to carry out this work.

Further soil geochem and geophysical surveys will be necessary to establish the presence of a porphyry Cu-Au system 1 km west of the Bullion Grid area as suggested by results of some of the 1991 work. High gold values in stringers obtained southeast of the Crackle Zone warrant follow-up. More work is necessary to expand and define the copper soil anomaly on the Willoughby Creek slope, south of the Crackle Zone. Elsewhere on the property, on the Humdinger Zone, it may be useful to run a few soil lines downslope of the Humdinger Zone prior to resuming trenching. Trenching should also be done uphill, to the south, to determine whether the zone continues in that direction.

Respectfully submitted,



D. Cremonese, P.Eng.
January 31, 1992

APPENDIX I -- WORK COST STATEMENT

Field Personnel--Period July 1 to Oct. 7, 1991:

E. Kruchkowski, Senior Geologist 10 days @ \$300/day	\$ 3,000
B. Game, Senior Geologist 27 days @ \$275/day	7,425
J. Nicholson, Geologist 27 days @ \$250/day	6,750
K. May, Geologist 27 days @ \$200/day	5,400
M. Boulding, Geological Technician 36 days @ \$200/day	7,200
T. Gustafson, Geological Assistant 10 days @ \$150/day	1,500
T. Archibald, Geological Assistant 10 days @ \$150/day	1,500
D. Cremoneese, P.Eng. 3 days @ \$350/day	1,050

Helicopter -- Highland Hel. (Ellsworth Logging Base)

Crew drop-offs/pick-ups, Camp-supplies mob/demob 37.2 hrs @ \$684.50	25,463
---	--------

Food -- 150 man-days @ \$25/man-day

3,750

**Mob/demob: Personnel/Equip./Supplies/Samples
(prorated with other projects where applicable)**

7,571

Field Supplies--camp lumber, consumables, etc.

1,650

Camp Rental (from contractor)

500

Truck Rental Charges

1,385

Equipment Rental (Pump, rock saw, plugger, generator)

1,812

Radio/Communications (includes hand-holds)

1,524

Stewart Accommodation

390

Expediting

1,328

Explosives & Related Supplies

918

Assays--Eco-Tech Labs, Kamloops, B.C.

Geochem Au, I.C.P. and rock sample preparation

349 @ \$16.00 sample

5,584

Geochem Au, I.C.P. and soil sample preparation

579 @ \$13.25 sample

7,672

Au/Ag/Cu/Pb/Zn/As Assays, Metallics, Disks, Etc.

1,101

Contractor's Office Charges (Nicholson & Assoc.)
(includes accounting, field map/report prep, etc.) 2,000

Report Costs

Report and map preparation, compilation and research	
D. Cremonese, P.Eng., 6 days @ \$350/day	2,100
Draughting -- RPM Computer	1,700
Word Processor - 10 hrs. @ \$25/hr.	250
Copies, report, jackets, maps, etc.	120
TOTAL.....	\$ 100,643

APPENDIX II - CERTIFICATE

I, Dino M. Cremonese, do hereby certify that:

1. I am a mineral property consultant with an office at Suite 509-675 W. Hastings, Vancouver, B.C.
2. I am a graduate of the University of British Columbia (B.A.Sc. in metallurgical engineering, 1972, and L.L.B., 1979).
3. I am a Professional Engineer registered with the Association of Professional Engineers of the Province of British Columbia as a resident member, #13876.
4. I have practiced my profession since 1979.
5. This report is based upon work carried out on the Croesus 1-4 and Bond 1-2 mineral claims, Skeena Mining Division from July to October of 1991. Reference to field notes and maps made by geologists Brian Game, J. Nicholson, Kevin May and Ed Kruchkowski is acknowledged. I have full confidence in the abilities of all samplers used in the 1991 geochemical program and am satisfied that all samples were taken properly and with care.
6. I am a principal of Teuton Resources Corp., owner of the Croesus 1-4 claims: this report was prepared solely for satisfying assessment work requirements in accordance with government regulations.

Dated at Vancouver, B.C. this 31st day of January, 1992.



D. Cremonese, P.Eng.

APPENDIX III

ASSAY CERTIFICATES

ECO-TECH LABORATORIES LTD.

10041 EAST TRANS CANADA HWY.
KAMLOOPS, B.C. V2C 2J3
PHONE - 604-573-5700
FAX - 604-573-4557

JULY 29, 1991

TEUTON RESOURCES CORP. - ETK 91-9003

602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1W2

VALUES IN PPM UNLESS OTHERWISE REPORTED

PAGE 1

PROJECT: DEL NORTH
219 SOIL SAMPLES RECEIVED JULY 17, 1991

RT#	DESCRIPTION	AV(ppb)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU PR(%)	E(%)	LA NG(%)	MN	MO MA(%)	NI	P	PB	SB	SN	SR Ti(%)	U	V	T	Zn							
1	0+00 B 0+25 B	260	1.6	1.22	35	1	650	(5	.32	9	33	3	517	6.34	.11	10	.61	6807	6	<.01	2	2050	1850	10	<20	39	.02	10	29	<10	4	1425
2	0+00 B 0+50 B	370	5.0	.53	145	2	215	(5	.62	32	86	(1	1340	>15	.01	<10	.48	8254	14	<.01	2	3260	172	40	<20	43	.01	30	6	<10	<1	3270
3	0+00 B 2+00 B	(5	<.2	1.44	40	1	65	(5	.16	(1	14	2	76	5.33	.11	<10	.59	774	(1	<.01	4	890	30	5	<20	10	.05	10	85	<10	<1	139
4	0+50 B 0+25 B	115	1.0	.60	15	6	305	(5	1.53	13	15	1	528	2.66	.09	<10	.36	5630	4	<.01	2	1670	109	5	20	25	.01	20	18	<10	3	942
5	0+50 B 0+50 B	10	<.2	2.13	40	1	210	(5	.46	1	22	4	97	4.01	.17	<10	.01	2396	1	<.01	7	1880	36	5	<20	25	.03	10	76	<10	4	100
6	0+50 B 0+75 B	(5	<.2	2.87	5	1	90	(5	.20	(1	20	4	132	4.32	.20	<10	.99	1247	(1	<.01	4	1070	16	5	<20	10	.08	10	95	<10	6	101
7	0+50 B 1+00 B	(5	<.2	1.44	65	10	230	(5	2.73	12	8	1	85	1.59	.06	<10	.43	2092	(1	<.01	5	1340	12	5	<20	143	.01	<10	31	<10	9	353
8	0+50 B 1+25 B	(5	<.6	.13	65	10	45	(5	.47	(1	1	(1	31	.31	.04	<10	.07	129	(1	<.01	(1	800	6	(5	<20	25	.01	10	4	<10	<1	60
9	0+50 B 1+50 B	(5	2.0	1.30	30	10	160	(5	.33	(1	31	7	40	5.90	.10	<10	>10000	1	<.01	4	4010	60	5	<20	26	.01	20	104	<10	<1	128	
10	0+50 B 1+75 B	45	<.2	1.17	30	2	155	(5	.49	(1	12	2	52	4.09	.00	<10	.33	657	1	<.01	2	840	30	5	<20	37	.03	<10	90	<10	1	86
11	0+50 B 2+25 B	10	2.0	1.37	55	6	65	(5	.20	(1	17	8	54	9.15	.03	<10	.23	1339	(1	<.01	10	1860	72	(5	<20	16	.01	20	44	<10	<1	73
12	0+50 B 2+50 B	35	1.0	1.50	25	2	70	(5	.34	(1	22	3	43	4.52	.04	<10	.20	3152	2	<.01	3	1910	36	(5	<20	30	.01	10	42	<10	4	82
13	0+50 B 2+75 B	40	<.2	2.57	40	2	90	(5	.32	(1	20	3	41	7.32	.23	<10	.75	1459	2	<.01	7	1650	48	5	<20	15	.04	10	100	<10	<1	113
14	0+50 B 3+00 B	155	3.2	1.20	95	4	140	(5	.41	1	21	2	56	5.25	.11	<10	.49	2715	1	<.01	6	1710	93	5	<20	25	.01	<10	37	<10	6	225
15	0+50 B 3+25 B	195	5.2	1.12	110	4	155	(5	.51	1	23	2	60	5.21	.07	<10	.46	2152	2	<.01	10	1750	162	10	<20	31	.01	<10	32	<10	8	251
16	0+50 B 3+50 B	200	6.0	1.11	125	4	160	(5	.61	1	23	(1	50	5.04	.07	<10	.47	2214	1	<.01	9	1730	116	5	<20	29	.01	<10	29	<10	8	256
17	0+50 B 4+00 B	155	3.6	1.65	220	2	120	(5	.29	4	49	4	83	6.03	.01	<10	.51	8054	8	<.01	36	1960	78	10	<20	14	.01	10	26	<10	12	355
18	1+00 B 0+25 B	25	<.2	1.14	50	2	50	(5	.04	(1	11	2	49	5.07	.12	<10	.38	499	(1	<.01	3	960	24	(5	<20	3	.06	10	107	<10	<1	72
19	1+00 B 0+50 B	15	<.2	.99	40	1	50	(5	.06	(1	10	2	45	5.55	.11	<10	.29	464	(1	<.01	2	1270	32	5	<20	3	.04	10	115	<10	<1	67
20	1+00 B 0+75 B	10	1.0	2.00	45	1	185	(5	1.16	4	20	1	84	4.37	.10	<10	.75	1373	1	<.01	4	1050	40	5	<20	70	.04	10	73	<10	1	255
21	1+00 B 1+00 B	20	<.2	.62	35	1	50	(5	.15	(1	6	(1	40	2.94	.05	<10	.12	225	1	<.01	2	540	24	(5	<20	14	.02	<10	91	<10	<1	70
22	1+00 B 1+25 B	25	<.4	2.81	15	10	200	(5	.71	3	20	5	70	4.35	.16	<10	.07	2222	1	<.01	5	1520	32	5	<20	47	.04	10	80	<10	12	211
23	1+00 B 1+50 B	(5	<.2	.79	15	20	40	(5	.09	(1	5	3	49	3.00	.04	<10	.10	289	(1	<.01	1	2710	18	(5	<20	5	.01	10	45	<10	<1	61
24	1+00 B 1+75 B	20	1.0	0.40	15	1	120	(5	.50	1	27	7	50	6.16	.06	<10	.36	3941	1	<.01	1	1610	62	(5	<20	35	.04	10	56	<10	30	90
25	1+00 B 2+00 B	25	<.2	1.03	40	1	200	(5	.02	(1	12	10	55	5.08	.03	<10	.24	1137	2	<.01	5	1460	24	5	<20	70	.02	10	92	<10	6	130
26	1+00 B 2+25 B	15	<.2	.57	300	6	60	(5	.06	(1	13	1	30	4.01	.01	<10	.12	2666	4	<.01	2	1420	66	(5	<20	6	.01	10	51	<10	<1	93

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TEUTON RESOURCES CORP. - ETK 91-9003

PAGE 7

BB#	DESCRIPTION	AU(DD)	AG AL(%)	AS	B	BA	BL CAl(%)	CD	CO	CR	CUPB(%)	E(%)	LA GNG(%)	MN	NO HAI(%)	NI	P	PB	SB	SH	SR Ti(%)	U	V	W	X	Z				
203	DHHS	12	135	3.8	2.98	145	2	60	CS .04	CL 15	6	49	5.44	.05	CL 10	.43	501	1	.01	0	520	114	5	C20	6	.01	CL 10	42	CL 100	
204	DHHS	13	35	.8	2.47	120	2	70	CS .03	CL 14	5	57	6.78	.00	CL 10	.49	773	1	.01	7	1110	100	5	C20	8	.01	CL 10	60	CL 100	
205	DHHS	14	*	100	0.2	1.09	100	2	150	CS .00	CL 26	1	39	4.65	.02	CL 10	.14	4445	1	.01	6	1250	92	5	C20	9	.01	CL 10	43	CL 200
206	DHHS	15	*	170	1.4	.00	175	2	65	CS .06	CL 11	CL 1	44	4.55	.04	CL 10	.13	1510	2	.01	6	1120	50	5	C20	9	.01	CL 10	53	CL 146
207	DHHS	16	220	2.0	1.02	125	2	115	CS .06	CL 10	3	53	5.49	.04	CL 10	.36	2520	1	.01	7	1380	84	5	C20	9	.01	CL 10	60	CL 143	
208	DHHS	17	870	7.6	1.52	315	2	60	CS .02	CL 11	CL 1	62	5.83	<.01	CL 10	.23	859	1	.01	5	860	112	5	C20	6	.01	CL 10	54	CL 230	
209	DHHS	18	*	50	.6	1.44	95	2	70	CS .00	CL 11	2	45	4.60	.09	CL 10	.35	1110	1	.01	4	1070	36	5	C20	11	.03	CL 10	77	CL 101
210	DHHS	19	20	0.2	2.93	100	4	90	CS .06	CL 19	6	87	6.21	.13	CL 10	.03	1400	CL .01	0	940	36	10	C20	8	.00	CL 10	109	CL 118		
211	DHHS	20	60	.8	2.53	90	2	60	CS .07	CL 12	3	90	5.39	.11	CL 10	.66	793	CL .02	5	910	16	5	C20	8	.03	CL 10	106	CL 97		
212	DHHS	21	CS	.8	2.57	65	2	70	CS .09	CL 15	3	95	5.23	.09	CL 10	.00	635	CL .01	6	830	20	5	C20	8	.06	CL 10	115	CL 84		
213	DHHS	22	CS	1.0	1.86	55	4	70	CS .19	CL 12	2	73	4.68	.00	CL 10	.63	687	CL .01	8	1200	10	5	C20	15	.04	CL 10	82	CL 85		
214	DHHS	23	10	.2	1.22	55	2	60	CS .00	CL 7	1	50	3.93	.06	CL 10	.23	403	CL .01	4	1000	10	5	C20	10	.02	CL 10	93	CL 79		
215	DHHS	24	20	.8	1.12	95	2	75	CS .09	CL 13	5	45	8.25	.05	CL 10	.19	1435	CL .01	4	1450	32	5	C20	9	.01	CL 10	45	CL 112		
216	DHHS	25	CS	0.2	2.69	90	2	75	CS .04	CL 13	5	45	8.25	.05	CL 10	.65	761	CL .01	6	820	30	5	C20	7	.09	CL 10	143	CL 90		
217	DHHS	26	45	.2	.82	60	2	60	CS .13	CL 4	CL 1	51	2.28	.01	CL 10	.09	234	1	.01	4	800	10	5	C20	14	.01	CL 10	62	CL 90	
218	DHHS	27	10	0.2	.70	70	4	45	CS .06	CL 5	2	45	2.01	.03	CL 10	.12	261	1	.01	3	750	12	5	C20	9	.01	CL 10	60	CL 77	
219	***	130	5.4	1.35	190	4	100	CS .20	3	35	4	63	5.23	<.01	CL 10	.48	5485	5	.01	32	1660	110	10	C20	20	.01	CL 10	20	CL 319	

NOTE: < = LESS THAN

* = -42 MESH

** = -20 MESH

*** = NO SAMPLE DESCRIPTION

CC: NICHOLSON ASSOC. INC.
606-675 WEST HASTINGS STREET
VANCOUVER, B.C.

Clint Ayers

ECO-TECH LABORATORIES LTD.
CLINT AYERS
LABORATORY MANAGER

ECO-TECH LABORATORIES LTD.

TEUTON RESOURCES CORP. - ETK 91-507

10041 EAST TRANS CANADA HWY.
 KAMLOOPS, B.C. V2C 2J3
 PHONE - 604-573-5700
 FAX - 604-573-4557

AUGUST 2, 1991

602 - 675 WEST HASTINGS STREET
 VANCOUVER, B.C.
 V6B 1N2

VALUES IN PPM UNLESS OTHERWISE REPORTED

PROJECT: DBL NORTH
 216 SOIL SAMPLES RECEIVED JULY 24, 1991

RT#	DESCRIPTION	ALU(pbb)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU PB(%)	E(%)	LA MG(%)	MN	NO MA(%)	NI	P	PB	SB	SN	SR TI(%)	U	V	W	X	Z							
1 - 6	+75 B	1 + 500	70	.8	1.86	120	6	60	(5	.03	<1	19	12	33	7.71	.05	10	.25	3252	9	.02	10	2100	82	5	<20	5	.04	<10	43	<10	<1	175
2 - 6	+75 B	1 +75 B	<5	2.0	2.26	50	4	110	(5	.02	<1	14	4	33	5.54	.13	<10	.36	2472	1	<.01	3	1300	46	5	<20	4	.01	<10	62	<10	<1	117
3 - 6	+75 B	2 +00 B	<5	.8	1.30	40	4	205	(5	.10	<1	16	4	26	6.02	.10	<10	.20	4368	1	.01	2	1730	66	5	<20	10	.01	<10	53	<10	<1	148
4 - 6	+75 B	2 +25 B	15	1.2	1.51	60	4	300	(5	.09	1	23	5	36	5.64	.13	<10	.25	9591	3	<.01	4	1460	68	5	<20	9	.01	<10	61	<10	<1	174
5 - 6	+75 B	2 +50 B	<5	1.4	1.07	55	4	500	(5	.01	<1	20	5	33	4.72	.10	<10	.19	9526	2	<.01	3	2120	100	5	<20	25	<.01	<10	40	<10	<1	199
6 - 6	+75 B	2 +75 B	5	.6	1.75	45	4	195	(5	.16	<1	19	4	39	5.25	.23	10	.46	4313	1	<.01	3	2950	56	5	<20	14	.01	<10	47	<10	<1	160
7 - 6	+75 B	3 +00 B	30	.8	1.71	20	4	110	(5	.06	<1	13	2	44	4.20	.18	10	.43	2941	<1	<.01	1	1340	36	5	<20	5	.01	<10	40	<10	<1	198
8 - 6	+75 B	3 +25 B	25	3.2	2.05	85	4	130	(5	.06	<1	17	4	62	4.99	.23	<10	.66	2592	<1	<.01	4	1890	32	5	<20	7	.02	<10	67	<10	<1	179
9 - 6	+75 B	3 +50 B	<5	6.2	1.30	25	4	135	(5	.19	<1	10	2	40	3.39	.10	10	.52	1270	<1	<.01	3	1300	34	5	<20	10	.01	<10	40	<10	<1	122
10 - 6	+75 B	3 +75 B	15	.4	1.96	35	4	340	(5	.22	<1	12	3	51	3.75	.13	10	.54	2191	<1	<.01	3	1290	44	5	<20	15	<.01	<10	43	<10	<1	174
11 - 6	+75 B	4 +00 B	40	.6	1.12	35	2	265	(5	.10	1	12	1	34	3.35	.11	10	.23	5026	<1	<.01	1	1300	26	5	<20	8	<.01	<10	39	<10	<1	112
12 - 7	+50 B	1 +00 B	45	.6	1.04	30	4	220	(5	.31	1	14	1	50	4.48	.11	10	.35	1947	<1	<.01	5	1200	32	5	<20	13	<.01	<10	20	<10	<1	179
13 - 7	+50 B	1 +25 B	60	1.0	1.20	45	4	225	(5	.38	1	19	3	69	5.30	.11	10	.44	2726	<1	<.01	6	1470	48	5	<20	19	<.01	<10	20	<10	<1	4
14 - 7	+50 B	1 +50 B	35	.6	.72	30	4	160	(5	.29	1	13	<1	50	4.30	.11	10	.20	1950	<1	<.01	2	1360	44	5	<20	10	.02	<10	36	<10	<1	172
15 - 7	+50 B	1 +75 B	200	1.0	.86	30	6	200	(5	.34	2	16	<1	75	4.97	.13	10	.30	2265	1	<.01	2	1410	56	5	<20	17	.02	<10	40	<10	<1	327
16 - 7	+50 B	2 +00 B	205	1.2	1.42	95	4	40	(5	.02	<1	16	13	20	5.66	.02	10	.15	2703	7	<.01	9	1230	88	5	<20	4	<.01	<10	20	<10	<1	180
17 - 7	+50 B	2 +25 B	10	1.2	1.61	50	2	165	(5	.03	1	10	2	25	4.37	.10	<10	.24	2753	<1	<.01	2	1240	32	5	<20	4	<.01	<10	46	<10	<1	133
18 - 7	+50 B	2 +50 B	15	2.0	1.93	60	2	140	(5	.03	1	14	4	31	4.60	.11	<10	.24	4292	<1	<.01	2	1570	54	5	<20	4	<.01	<10	50	<10	<1	160
19 - 7	+50 B	2 +75 B	25	1.6	1.94	130	4	210	(5	.15	<1	15	3	61	4.53	.19	10	.50	2372	<1	<.01	4	1350	68	10	<20	14	<.01	<10	52	<10	<1	219
20 - 7	+50 B	3 +00 B	20	1.0	2.43	40	4	120	(5	.06	<1	16	4	58	4.00	.17	10	.56	2020	<1	<.01	2	1310	24	5	<20	5	.02	<10	59	<10	<1	176
21 - 7	+50 B	3 +25 B	115	1.0	1.40	25	4	100	(5	.29	1	15	1	105	4.29	.16	20	.38	2453	<1	<.01	2	1690	26	5	<20	13	.04	<10	44	<10	<1	220
22 - 7	+50 B	3 +50 B	85	.4	2.00	25	2	530	(5	.36	<1	11	3	67	3.68	.10	10	.37	2757	<1	<.01	2	1940	28	5	<20	22	<.01	<10	35	<10	<1	250
23 - 7	+50 B	4 +00 B	40	.8	.89	25	4	420	(5	.61	2	15	1	41	2.86	.12	<10	.19	6691	<1	<.01	1	2320	60	5	<20	24	<.01	<10	23	<10	<1	180
24 - 8	+00 B	0 +25 B	60	1.0	1.60	105	2	150	(5	.12	<1	23	1	34	5.75	.07	10	.22	3475	2	<.01	6	1530	90	5	<20	8	<.01	<10	28	<10	<1	190
25 - 8	+00 B	0 +50 B	55	1.6	1.31	135	4	210	(5	.17	<1	20	2	25	5.01	.05	10	.16	3732	1	<.01	9	2400	102	5	<20	12	<.01	<10	21	<10	<1	198
26 - 8	+00 B	0 +75 B	35	1.0	1.30	100	4	255	(5	.03	1	24	1	19	6.20	.09	10	.15	7112	3	<.01	5	2140	48	5	<20	7	<.01	<10	27	<10	<1	176

ECO-TECH LABORATORIES LTD.

TSUTON RESOURCES CORP. - ETK 91-507

AGE 7

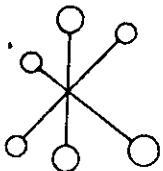
#	DESCRIPTION	AU(dob)	AG AL(%)	AS	B	BA	BI CN(%)	CD	CO	CR	CU PB(%)	E(%)	LA MG(%)	MN	MO MA(%)	MI	P	PB	SB	SM	SR TI(%)	U	V	W	X	Z
203-L12 +00	2 +25	25	1.4 1.25	90	4	105	<5 .09	<1	32	14	36 4.78	.02	50 .31	3922	1 <.01	54	1600	20	10 <20	9 <.01	<10	15 <10	1	135		
204-L12 +00	2 +50	40	.6 1.01	80	4	95	<5 .09	<1	34	18	33 4.86	.03	50 .40	3562	1 <.01	53	1800	16	10 <20	10 <.01	<10	10 <10	2	128		
205-L12 +00	2 +75	50	.6 1.00	80	4	120	<5 .20	<1	39	22	40 4.94	.03	50 .50	4223	2 <.01	55	2000	22	10 <20	17 <.01	<10	21 <10	2	134		
206-L12 +00	3 +00	55	.6 1.79	75	4	115	<5 .07	<1	51	27	36 5.11	.02	40 .48	5027	1 <.01	53	1910	20	15 <20	9 <.01	<10	21 <10	<1	141		
207-L12 +00	0 +00	30	.6 1.44	155	6	115	<5 .13	1	46	39	55 5.26	<.01	50 .77	2093	2 <.01	101	1090	26	15 <20	17 <.01	<10	17 <10	1	147		
208-L12 +00	0 +25	45	1.0 2.21	65	4	335	<5 .15	<1	24	1	24 4.60	.05	40 .71	7472	1 <.01	3	1120	6	10 <20	21 <.01	<10	29 <10	6	124		
209-L12 +00	0 +50	40	.4 1.33	80	4	155	<5 .07	<1	15	2	13 3.93	.00	30 .19	3418	2 <.01	4	2180	26	5 <20	11 <.01	<10	29 <10	<1	142		
210-L12 +00	B.L.0 +75	45	.2 1.42	95	4	110	<5 .06	<1	10	1	14 3.67	.04	30 .18	1860	2 <.01	3	2310	20	5 <20	8 <.01	<10	19 <10	<1	126		
211-L12 +00	1 +00	65	.4 2.54	80	4	80	<5 .06	<1	13	7	14 3.66	.04	30 .19	2227	1 <.01	2	2400	06	5 <20	5 <.01	<10	28 <10	<1	161		
212-L12 +00	1 +25	60	1.2 1.99	95	4	110	<5 .08	<1	15	5	19 4.51	.04	40 .20	3116	1 <.01	4	2360	130	5 <20	7 <.01	<10	27 <10	<1	236		
213-L12 +00	1 +50	65	1.4 1.00	80	4	215	<5 .20	1	10	3	28 3.99	.07	40 .20	5014	1 <.01	4	2250	72	5 <20	18 <.01	<10	31 <10	9	206		
214-L12 +00	2 +25	60	.6 1.34	85	4	135	<5 .30	1	16	1	24 4.12	.06	40 .24	3717	2 <.01	4	2420	76	5 <20	29 <.01	<10	24 <10	5	203		
215-L12 +00	2 +50	105	3.0 .98	75	4	105	<5 .33	1	24	1	30 5.06	.04	50 .21	4084	12 <.01	27	1740	62	5 <20	24 <.01	<10	19 <10	16	237		
216-L12 +00	2 +75	35	3.4 1.23	170	4	140	<5 .22	1	40	6	67 7.01	<.01	60 .31	4409	4 <.01	18	1690	224	10 <20	18 <.01	<10	30 <10	12	297		

OTR: < = LESS THAN
* = TO -42 MESH

C.C.: NICHOLSON AND ASSOCIATES
606-675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1H2



ECO-TECH LABORATORIES LTD.
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ASSAYING - ENVIRONMENTAL TESTING
10041 East Trans Canada Hwy., Kamloops, B.C. V2C 2J3 (604) 573-5700 Fax 573-4557

AUGUST 6, 1991

CERTIFICATE OF ASSAY ETK 91-524

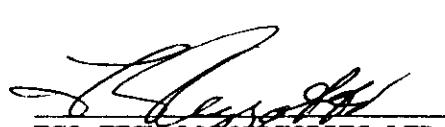
TEUTON RESOURCES CORP.
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1N2

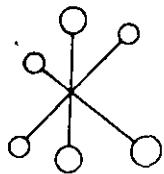
SAMPLE IDENTIFICATION: 161 ROCK samples received JULY 25, 1991
----- PROJECT: DEL NORTE

ET#	Description	AU (g/t)	AU (oz/t)	AG (g/t)	AG (oz/t)	PB (%)	ZN (%)
7	82407	1.00	.03	-	-	-	-
15	82415	25.63 *	.75	1790.	52.21	-	-
16	82416	-	-	31.5	.92	-	-
18	82418	-	-	30.8	.90	-	-
19	82419	-	-	429.9	12.54	2.08	-
29	82502	9.12 *	.27	1069.	31.18	4.40	-
31	82504	11.81 *	.34	51.9	1.51	-	1.46
39	82512	7.71 *	.23	181.4	5.29	-	-
44	82517	-	-	48.1	1.40	-	-
45	82518	-	-	70.8	2.07	1.24	1.16
46	82519	1.62	.05	-	-	-	-
48	82521	-	-	30.1	.88	-	-
49	82646	17.78 *	.52	1245.	36.31	4.68	9.94
76	81776	-	-	-	-	-	1.48
83	81783	-	-	41.2	1.20	-	-
86	81786	-	-	35.8	1.04	-	-
87	81787	-	-	26.8	.78	-	-
101	81801	-	-	27.9	.81	-	-
125	82623	1.26	.04	-	-	-	-
133	82631	-	-	-	-	1.96	1.18
135	82633	-	-	112.1	3.27	-	-
141	82639	2.81	.08	58.1	1.69	-	-
144	82642	1.86	.05	52.1	1.52	-	-
148	82646	10.75 *	.31	571.7	16.67	-	-
157	82655	30.00 *	.88	3407.	99.37	10.10	-
158	82656	4.30	.13	335.0	9.77	-	-
159	82657	4.74	.14	827.8	24.14	1.18	-
160	82658	2.84	.08	145.2	4.23	-	-
161	82659	-	-	47.1	1.37	-	-

NOTE: * = SAMPLE SCREENED AND METALLIC ASSAYED

CC: NICHOLSON & ASSOCIATES
VANCOUVER, B.C.


ECO-TECH LABORATORIES LTD.
FRANK J. PEZZOTTI
B.C. Certified Assayer



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METALLIC CALCULATION

SAMPLE NUMBER	-140 VALUE	+140 VALUE	CALCULATED VALUE
524-15	18.33	149.7714	25.62934
524-29	9.189999	8.042028	9.121501
524-31	12.03	7.778765	11.80721
524-39	7.72	7.542723	7.714463
524-49	18.16	12.43001	17.77796
524-148	12.81	3.690645	10.75405
524-157	27.95	51.10697	30.00473
524-158	2.31	767.0347	4.29964
524-159	5.12	3.429408	4.743171

ECO-TECH LABORATORIES LTD.

10041 EAST TRANS CANADA HWY.
KAMLOOPS, B.C. V2C 2J3
PHONE - 604-573-5700
FAX - 604-573-4557

AUGUST 1, 1991

TEUTON RESOURCES CORP. - ETK 91-524

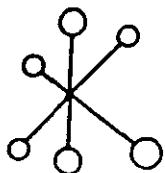
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1H2

VALUES IN PPM UNLESS OTHERWISE REPORTED

PAGE 1

PROJECT: DBL NORTH
161 ROCK SAMPLES RECEIVED JULY 25, 1991

ETK	DESCRIPTION	AU(ppm)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU PB(%)	K(%)	LA MG(%)	MN	NO NA(%)	NI	P	PB	SB	SM	SR Ti(%)	V	V	V	T	Zn
1 -	02401	30 .4 1.00	50	4 90	<5	2.12	<1	9	33	13	3.27	.17	10	.29	1220	1 <.01	3	1220	20	<20	27 <.01	<10	12 <10	1 73		
2 -	02402	25 1.0 .63	500	6 100	<5	.72	<1	30	14	65	5.61	<.01	<10	.15	1340	1 <.01	32	1140	18	<20	20 <.01	<10	11 <10	<1 152		
3 -	02403	5 1.8 .51	205	4 65	<5	.55	<1	23	15	63	5.53	.14	<10	.13	850	1 <.01	24	1060	16	<20	24 <.01	<10	8 <10	<1 132		
4 -	02404	<5 .4 .64	25	4 125	<5	2.06	<1	9	37	12	3.27	.20	10	.21	1550	2 <.01	1	1310	6	<20	51 <.01	<10	7 <10	<1 102		
5 -	02405	<5 <.2 .49	15	4 95	<5	5.74	<1	6	29	7	3.16	.19	<10	.96	2217	1 <.01	<1	1550	2	<20	225 <.01	<10	13 <10	<1 52		
6 -	02406	10 1.0 .29	265	6 50	<5	.05	<1	14	37	9	9.17	.05	<10	.14	95	21 <.01	<1	950	126	25 <20	10 <.01	<10	<1 <10	<1 55		
7 -	02407	>1000 7.2 .12	165	6 30	<5	1.15	<1	7	172	28	2.50	<.01	<10	.14	2036	10 <.01	7	290	46	10 <20	24 <.01	<10	3 <10	<1 77		
8 -	02408	20 .2 .47	20	4 75	<5	3.52	<1	11	40	47	3.39	.20	10	.44	1950	3 <.01	<1	1260	16	<20	113 <.01	<10	14 <10	<1 119		
9 -	02409	<5 .4 .52	35	4 50	<5	2.42	<1	15	43	41	4.32	.23	<10	.30	1935	3 <.01	<1	1430	56	<20	53 <.01	<10	16 <10	<1 149		
10 -	02410	10 1.0 .43	30	4 40	<5	2.15	2	15	45	83	3.07	.21	<10	.24	1975	4 <.01	1	1300	322	<20	56 <.01	<10	9 <10	<1 338		
11 -	02411	<5 2.2 .20	20	6 50	<5	12.13	2	8	40	6	2.50	.09	<10	1.02	1656	40 <.01	<1	490	94	10 <20	870 <.01	<10	3 <10	<1 375		
12 -	02412	<5 1.0 .41	40	4 170	<5	2.65	1	10	50	4	1.90	.16	<10	.24	1145	7 <.01	<1	1120	138	<20	126 <.01	<10	3 <10	<1 304		
13 -	02413	160 .0 .40	55	4 70	<5	2.54	2	19	56	55	4.84	.11	<10	.31	1583	12 <.01	15	1070	26	<20	58 <.01	<10	16 <10	<1 188		
14 -	02414	300 .4 .47	770	6 55	<5	4.61	<1	11	26	5	4.89	<.01	<10	.76	1383	1 <.01	<1	1650	<2	<20	210 <.01	<10	5 <10	<1 45		
15 -	02415	>1000 >30 .05	55	2 10	10	.07	15	1	102	1979	.59	<.01	<10	.02	370	14 <.01	1	40	4636	1400 <20	8 <.01	<10	1 <10	<1 406		
16 -	02416	230 >30 .44	135	4 30	<5	.06	<1	4	78	88	3.17	.07	10	.05	145	19 <.01	15	1110	710	40 <20	19 <.01	<10	18 <10	<1 488		
17 -	02417	30 9.6 1.20	105	6 85	<5	1.63	1	4	36	57	3.87	.27	10	.11	110	16 <.01	13	>10000	70	15 <20	138 <.01	<10	17 <10	<1 373		
18 -	02418	430 >30 .22	190	4 20	<5	.06	5	3	144	39	2.05	.03	10	.05	301	13 <.01	6	420	756	30 <20	9 <.01	<10	8 <10	<1 472		
19 -	02419	605 >30 .12	215	2 15	<5	.02	149	1	253	219	1.74	<.01	<10	.02	56	17 <.02	3	250	>10000	370 <20	10 <.01	<10	3 <10	<1 9331		
20 -	02420	205 17.8 .19	140	4 15	<5	.37	9	3	165	25	2.36	.03	<10	.12	444	12 <.01	7	440	326	15 <20	37 <.01	<10	5 <10	<1 608		
21 -	02421	140 3.6 .54	30	2 130	<5	.04	7	14	130	50	4.51	.23	10	.16	5294	8 <.01	1	990	1588	<20	5 <.01	<10	11 <10	<1 1641		
22 -	02422	115 3.6 .29	20	4 290	<5	.03	13	19	99	48	2.62	.16	10	.06	6111	8 <.01	1	580	1042	<20	8 <.01	<10	6 <10	<1 1786		
23 -	02423	35 1.2 .37	10	4 70	<5	.04	2	6	100	29	1.99	.24	10	.05	1411	5 <.01	<1	570	980	<20	3 <.01	<10	6 <10	<1 495		
24 -	02424	45 2.4 .38	15	4 60	<5	.05	3	11	135	45	2.92	.14	10	.14	2434	8 <.01	51	7 <.01	3	40	138	175 <20	7 <.01	<10	<1 729	
25 -	02425	840 14.0 .04	5655	6 15	<5	.01	<1	14	176	336	2.62	<.01	<10	.03	51	7 <.01	3	40	138	175 <20	7 <.01	<10	<1 49			
26 -	02426	100 17.4 .42	330	6 65	<5	.32	4	26	49	109	5.10	.09	10	.10	1472	4 <.01	43	1700	246	20 <20	19 <.01	<10	10 <10	<1 390		



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AUGUST 6, 1991

CERTIFICATE OF ASSAY ETK 91-563

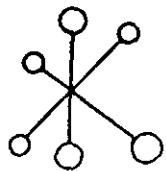
TEUTON RESOURCES CORP.
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1N2

SAMPLE IDENTIFICATION: 28 ROCK samples received JULY 31, 1991

PROJECT: DEL NORTE

ET#	Description	AU (g/t)	AU (oz/t)	AG (g/t)	AG (oz/t)	CU (%)	AS (%)
1	-	82428	-	-	-	-	-
2	-	82429	-	-	30.3	.88	-
3	-	82522	-	-	-	-	-
4	-	82523	4.26	.124	-	-	-
5	-	82524	-	-	-	-	-
6	-	82525	5.61	.164	-	-	2.86
7	-	82526	2.93	.085	-	-	-
8	-	82527	-	-	-	-	-
9	-	82528	-	-	-	-	-
10	-	82529	-	-	-	-	-
11	-	82530	-	-	-	-	-
12	-	82531	14.52	.423	-	-	-
13	-	82532	22.33	.651	33.2	.97	-
14	-	82533	-	-	-	1.54	2.69
15	-	82534	-	-	-	-	-
16	-	82535	-	-	-	-	-
17	-	82536	2.00	.058	32.6	.95	-
18	-	DN-KM-R	81803	-	-	-	4.86
19	-	DN-KM-R	81804	-	-	-	-
20	-	DN-KM-R	81805	-	-	-	-
21	-	DN-KM-R	81806	-	-	-	-
22	-	DN-KM-R	81807	-	-	-	-
23	-	DN-KM-R	81808	-	-	-	-
24	-	DN-KM-R	81809	2.79	.081	-	1.00
25	-	DN-KM-R	81810	-	-	-	-
26	-	DN-KM-R	81811	-	-	-	-
27	-	DN-KM-R	81812	-	-	-	-
28	-	DN-KM-R	81813	-	-	-	-


ECO-TECH LABORATORIES LTD.
FRANK J. PEZZOTTI, A.Sc.T.
B.C. Certified Assayer



ECO-TECH LABORATORIES LTD.

ASSAYING - ENVIRONMENTAL TESTING

10041 East Trans Canada Hwy., Kamloops, B.C. V2C 2J3 (604) 573-5700 Fax 573-4557

METALLIC CALCULATION

SAMPLE NUMBER	-140 VALUE	+140 VALUE	CALCULATED VALUE
563-6	5.32	9.588169	5.607578
563-12	13.64	32.31123	14.51564
563-13	21.75	43.95413	22.32812

ECO-TECH LABORATORIES LTD.

TEUTON RESOURCES CORP. - ETR 91-567

10011 EAST TRAMS CANADA MILE.
KAMLOOPS, B.C. V2C 2J3
PHONE - 604-573-5700
FAX - 604-573-4557

602 - 675 WEST Hastings STREET
VANCOUVER, B.C.
V6B 1B2

AUGUST 9, 1991

VALUES IN PPM UNLESS OTHERWISE REPORTED

PROJECT: DEL NORTE
132 SOIL SAMPLES RECEIVED JULY 31, 1991

#	DESCRIPTION	AO(ppb)	Ag AL(%)	As	B	BA	Bi CA(%)	CD	Co	Cr	CU PB(%)	I(%)	LA BG(%)	M	Mo MA(%)	Si	P	Pb	Sb	SB	SR Ti(%)	U	V	V	T	U					
1 - DU BG - 5 - 01		20	.4	3.15	30	6	400	<5	.33	<1	38	4	279	8.73	.18	30	.91	4603	<1	<.01	10	2010	10	5	<20	16	.01	<10	15	112	
2 - DU BG - 5 - 02		30	.2	3.37	10	4	325	<5	.47	<1	25	3	184	7.23	.21	10	.63	3312	<1	<.01	8	2260	16	5	<20	21	.01	<10	100	<10	116
3 - DU BG - 5 - 03		15	.2	3.09	20	4	145	<5	.19	<1	27	7	184	7.19	.15	20	.57	2531	<1	<.01	9	2400	12	5	<20	8	.01	<10	76	<10	96
4 - DU BG - 5 - 04		10	.2	2.88	<5	6	120	<5	.18	<1	21	7	216	4.63	.18	10	.53	2048	<1	<.01	7	1990	8	5	<20	12	<.01	<10	67	<10	100
5 - DU BG - 5 - 05		25	.4	1.97	45	6	170	<5	.28	<1	36	1	223	8.91	.12	20	.76	4742	<1	<.01	11	1800	10	5	<20	14	.01	<10	68	<10	113
6 - DU BG - 5 - 06		10	.2	3.02	10	4	130	<5	.21	<1	22	1	183	6.66	.19	10	.66	2554	<1	<.01	7	2680	14	5	<20	10	.01	<10	68	<10	126
7 - DU BG - 5 - 07		5	.2	3.34	5	4	135	<5	.13	<1	22	3	152	5.74	.24	10	.64	2086	1	<.01	7	2450	10	5	<20	6	.01	<10	88	<10	99
8 - DU BG - 5 - 08		15	.2	3.69	<5	4	115	<5	.06	<1	21	5	145	5.66	.20	10	.66	1588	2	<.01	6	2110	10	5	<20	5	.01	<10	96	<10	104
9 - DU BG - 5 - 09		15	1.2	3.02	15	6	190	<5	.18	<1	29	3	236	6.89	.14	30	.79	>10000	1	<.01	8	1400	6	5	<20	10	<.01	<10	59	<10	78
10 - DU BG - 5 - 10		5	.2	1.76	20	6	120	<5	.22	<1	23	11	55	5.30	.10	10	.36	3456	1	<.01	4	2530	10	5	<20	13	.01	<10	47	<10	94
11 - DU BG - 5 - 11		5	1.0	3.30	35	6	285	<5	.24	<1	34	1	221	10.39	.13	30	.02	>10000	1	<.01	15	1810	2	5	<20	22	.01	<10	63	<10	73
12 - DU BG - 5 - 12		10	.2	3.59	<5	6	75	<5	.14	<1	19	13	131	5.57	.11	10	.59	1858	2	<.01	8	2120	14	5	<20	9	.01	<10	65	<10	97
13 - DU BG - 5 - 13		10	.2	3.93	<5	6	110	<5	.20	<1	25	5	193	5.01	.15	20	.03	2644	1	<.01	10	1600	14	5	<20	14	.01	<10	68	<10	100
14 - DU BG - 5 - 14		5	.2	3.80	<5	6	65	<5	.20	<1	22	11	140	5.06	.12	10	.63	1412	2	<.01	10	2150	14	5	<20	11	.01	<10	61	<10	95
15 - DU BG - 5 - 15		5	.4	2.69	25	4	155	<5	.21	<1	25	6	85	6.83	.11	10	.50	3173	1	<.01	9	2930	12	5	<20	13	.01	<10	84	<10	120
16 - DU BG - 5 - 16		5	.2	2.88	25	4	100	<5	.07	<1	21	8	98	6.37	.11	10	.45	2632	1	<.01	10	2020	12	5	<20	7	.01	<10	82	<10	99
17 - DU BG - 5 - 17		5	.2	4.65	<5	6	45	<5	.13	<1	13	25	86	5.67	.08	20	.41	923	3	.02	9	1770	24	5	<20	7	.03	<10	41	<10	81
18 - DU BG - 5 - 18		10	.2	2.89	10	6	60	<5	.10	<1	32	14	113	5.29	.11	10	.03	1873	2	<.01	21	1670	16	5	<20	9	.01	<10	62	<10	86
19 - DU BG - 5 - 19		<5	.4	2.59	25	6	130	<5	.09	<1	22	12	83	5.56	.12	10	.43	2464	2	<.01	15	2200	18	5	<20	8	<.01	<10	71	<10	97
20 - DU BG - 5 - 20		<5	.2	3.21	15	6	60	<5	.05	<1	19	21	76	6.01	.10	10	.53	1623	2	<.01	15	1920	18	5	<20	5	.01	<10	63	<10	93
21 - DU BG - 5 - 21		10	.2	2.31	20	4	90	<5	.07	<1	23	10	76	5.66	.13	10	.45	1515	1	<.01	17	2470	16	5	<20	6	<.01	<10	54	<10	99
22 - DU BG - 5 - 22		130	2.0	2.87	20	4	100	<5	.19	3	30	14	95	5.46	.12	10	.44	3029	1	<.01	23	2510	612	5	<20	11	.01	<10	38	<10	712
23 - DU BG - 5 - 23		15	.8	2.54	30	4	100	<5	.11	<1	29	13	91	5.41	.15	20	.49	5215	1	<.01	21	2910	58	5	<20	8	.01	<10	52	<10	201
24 - DU BG - 5 - 24		<5	.6	3.35	15	6	155	<5	.15	<1	21	13	82	5.29	.10	30	.40	2297	<1	<.01	17	1810	32	5	<20	12	.01	<10	39	<10	129
25 - DU BG - 5 - 25		<5	.6	2.29	20	6	100	<5	.10	<1	20	7	105	5.02	.11	20	.61	3596	1	<.01	19	1370	30	5	<20	10	.01	<10	40	<10	129
26 - DU BG - 5 - 26		<5	.2	2.17	25	4	120	<5	.11	<1	10	12	41	5.33	.10	10	.20	1167	1	<.01	7	3220	24	5	<20	8	.01	<10	57	<10	115

ECO-TECH LABORATORIES LTD.
10041 EAST TRANS CANADA HWY.
KAMLOOPS, B.C. V2C 2J3
PHONE - 604-573-5700
FAX - 604-573-4557

GUST 19, 1991

TRUTON RESOURCES CORP. - ETK 91-623
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1H2

VALUES IN PPM UNLESS OTHERWISE REPORTED

PROJECT: TWILIGHT
39 SOIL SAMPLES RECEIVED AUGUST 12, 1991

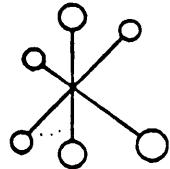
	DESCRIPTION	AU(ppb)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU FE(%)	K(%)	LA MG(%)	MN	MO NA(%)	NI	P	PB	SB	SN	SR TI(%)	U	V	W	X	ZN
1-	DNKMD 14	140	2.4 3.90	550	10	180	<5 .82	<1	43	11	376 3.22	.69	10 2.00	1300	2 .11	17 1520	26	30 <20	37 .18	<10 177	<10 15	15 172				
2-	DNKMD 15	55	2.8 3.03	230	12	150	<5 1.16	<1	41	12	270 3.20	.48	10 1.52	1304	2 .08	14 1890	26	15 <20	46 .11	<10 131	<10 11	11 161				
3-	DNKMD 16	30	1.2 3.28	130	12	110	<5 .82	<1	40	12	255 3.33	.34	10 1.41	1286	3 .07	14 2170	28	15 <20	31 .06	<10 120	<10 15	15 138				
4-	DNKMD 17	95	1.8 2.61	330	10	85	<5 .77	<1	30	12	181 3.34	.30	10 1.33	1310	2 .09	14 1940	26	15 <20	38 .09	<10 104	<10 13	12 128				
5-	DNKMD 18	80	2.4 2.35	380	10	80	<5 .69	<1	30	10	172 3.21	.31	10 1.28	1310	3 .07	12 1670	30	25 <20	33 .08	<10 95	<10 12	11 116				
6-	DNKMD 19	115	2.4 2.86	385	12	90	<5 .80	<1	34	11	202 3.20	.30	10 1.40	1308	2 .09	15 1990	28	25 <20	41 .11	<10 113	<10 13	13 134				
7-	DNKMD 20	50	1.2 3.69	230	10	190	<5 .61	<1	43	13	334 3.18	.66	10 2.25	1299	3 .08	14 1460	12	20 <20	30 .20	<10 177	<10 12	12 121				
8-	DNKMD 21	30	.6 4.77	65	8	120	<5 .19	<1	36	12	285 3.18	.61	<10 2.09	1300	5 .09	12 2290	10	15 <20	13 .16	<10 209	<10 6	145				
9-	DNKMD 22	25	.8 4.18	325	6	180	<5 1.31	<1	53	10	380 3.18	.92	<10 2.28	1300	3 .09	13 1720	10	15 <20	47 .22	<10 227	<10 12	12 149				
0-	DNKMD 23	10	.6 2.14	65	6	135	<5 .89	<1	16	7	58 3.22	.62	<10 .99	812	2 .06	6 2380	12	10 <20	44 .11	<10 162	<10 2	95				
1-	DNKMD 24	10	.6 4.30	65	6	85	<5 .25	<1	26	10	141 3.20	.40	10 1.35	1341	1 .07	9 1290	8	15 <20	17 .14	<10 168	<10 10	10 105				
2-	DNKMD 25	<5	.4 .82	20	14	75	<5 .56	<1	8	4	52 1.91	.20	<10 .32	544	1 .02	4 1660	2	5 <20	18 .05	<10 48	<10 2	96				
3-	DNKMD 26	<5	.8 3.75	30	12	115	<5 .15	<1	27	11	109 3.19	.44	<10 1.34	1304	2 .08	8 1310	8	15 <20	11 .17	<10 209	<10 6	84				
4-	DNKMD 27	25	1.2 5.05	90	8	145	<5 .33	<1	27	9	222 3.20	.64	10 1.66	1292	2 .07	11 1540	10	15 <20	20 .20	<10 165	<10 9	99				
5-	DNKMD 28	220	5.0 3.35	1180	4	60	<5 .39	<1	45	19	340 3.18	.40	10 1.35	1294	3 .10	20 1360	40	55 <20	17 .07	<10 129	<10 14	14 165				
6-	DNKMD 29	50	1.4 2.81	375	10	100	<5 .15	<1	40	11	186 3.21	.42	<10 .86	1241	2 .09	10 2180	22	20 <20	10 .06	<10 165	10	5 117				
7-	DNKMD 30	35	2.8 3.60	405	6	90	<5 .17	<1	67	22	489 3.18	.28	10 1.02	1134	4 .07	19 3780	22	20 <20	11 .05	<10 116	<10 9	136				
8-	DNKMD 31	130	4.0 5.68	320	6	100	<5 .11	<1	35	8	247 3.19	.55	10 1.67	1294	1 .08	10 1120	30	20 <20	13 .17	<10 169	<10 11	148				
9-	DNKMD 32	20	.2 4.09	160	6	70	<5 .19	<1	25	6	158 3.35	.18	<10 1.64	1129	3 .07	8 1020	10	15 <20	17 .15	<10 203	<10 4	88				
10-	DNKMD 33	130	4.2 5.01	245	10	190	<5 .39	2	55	7	1075 2.91	.28	<10 1.69	1131	3 .12	11 1690	28	25 <20	35 .15	<10 171	<10 14	150				
11-	DNKMD 34	85	1.4 4.54	160	14	205	<5 .97	1	44	7	503 2.92	1.16	<10 2.46	1144	5 .13	10 1500	8	15 <20	43 .26	<10 218	<10 19	126				
12-	DNKMD 35	5	.2 .14	10	10	85	<5 .57	1	2	1	21 .39	.05	<10 .05	220	<1 .01	2 770	6	<5 <20	25 .01	<10 6	<10 4	44				
13-	DNKMD 36	20	1.0 3.56	165	6	75	<5 .29	<1	39	6	310 2.93	.32	<10 1.57	1173	3 .10	8 1170	10	15 <20	18 .14	<10 169	10	5 128				
14-	DNKMD 37	30	.6 4.09	155	12	155	<5 .47	<1	69	12	399 2.91	.97	<10 2.16	1153	3 .12	12 1440	6	20 <20	19 .24	<10 260	10	8 131				
15-	DNKMS 97	40	3.2 3.88	345	8	155	<5 .09	<1	27	14	237 3.07	.46	<10 1.55	1170	4 .09	12 1190	16	20 <20	9 .10	<10 166	<10 6	131				
16-	DNKMS 98	5	1.2 2.28	125	8	50	<5 .03	<1	21	11	102 3.01	.20	<10 .82	1170	3 .09	7 1020	14	15 <20	5 .06	<10 144	<10 5	102				

E 2	TEUTON RESOURCES RTK 91-623											AUGUST 19, 1991											ECO-TECH LABORATORIES LTD.																												
	DESCRIPTION		AU(ppb)		AG AL(%)		AS		B		BA		BI CA(%)		CD		CO		CR		CU FE(%)		K(%)		LA MG(%)		Mn		Mo Na(%)		Ni		P		PB		SB		SW		Sr Ti(%)		U		V		W		Y		Zn
7-	DNMS	99	5	.2	1.34	120	8	30	<5	.11	<1	8	5	62	2.98	.19	<10	.38	359	2	.05	5	710	12	10	<20	9	.04	<10	187	<10	2	60																		
8-	DNMS	100	5	.8	2.55	70	6	70	<5	.07	<1	15	5	75	2.94	.19	<10	.83	1041	1	.07	5	1010	10	15	<20	11	.05	<10	130	<10	3	72																		
9-	DNMS	101	35	3.2	2.07	95	4	225	<5	.13	<1	28	9	84	2.95	.28	<10	.63	963	2	.06	5	2220	20	15	<20	12	.04	<10	126	<10	3	150																		
10-	DNMS	102	10	2.6	2.10	160	8	70	<5	.18	<1	49	8	89	2.93	.22	<10	.45	1039	4	.08	5	5610	20	15	<20	13	.02	<10	120	<10	4	104																		
11-	DNMS	103	5	1.0	1.05	105	6	25	<5	.10	<1	7	4	95	2.97	.09	<10	.16	608	3	.05	3	2660	10	10	<20	9	.01	<10	62	<10	3	59																		
12-	DNMS	104	10	1.2	4.46	115	10	170	<5	.24	<1	36	12	448	2.94	.84	<10	1.98	1156	3	.10	13	1170	10	20	<20	14	.19	<10	195	<10	8	113																		
13-	DNMS	105	5	1.0	3.85	70	12	185	<5	.31	<1	32	12	188	2.94	.43	<10	1.67	1135	3	.09	10	2040	8	15	<20	15	.16	<10	220	<10	2	152																		
14-	DNMS	106	<5	1.0	3.16	45	10	250	<5	.29	<1	31	11	169	2.94	.32	<10	1.39	1120	3	.08	8	930	10	15	<20	13	.22	<10	212	<10	2	93																		
15-	DNMS	107	5	1.0	3.55	70	10	85	<5	.20	<1	34	11	383	2.92	.55	<10	1.55	1125	4	.10	8	5370	8	15	<20	11	.12	<10	178	<10	6	100																		
16-	DNMS	108	<5	.4	2.87	65	10	140	<5	.47	1	27	7	106	2.94	.77	<10	1.25	1164	2	.08	8	980	12	15	<20	26	.11	<10	151	<10	6	115																		
17-	DNMS	109	<5	.8	3.89	10	14	220	<5	1.00	1	35	9	245	2.94	1.03	<10	1.96	1136	3	.13	10	2370	8	20	<20	54	.14	<10	183	<10	11	174																		
18-	DNMS	110	<5	.4	2.87	15	8	55	<5	.06	<1	22	10	87	2.91	.20	<10	1.30	686	2	.11	7	700	8	15	<20	6	.36	<10	279	<10	2	105																		
19-	DNMS	111	<5	.8	1.76	40	6	50	<5	.06	<1	8	5	86	2.98	.21	<10	.42	303	2	.04	3	1450	6	10	<20	5	.03	<10	76	<10	3	55																		

USE NOTE: AS ASSAYS TO FOLLOW

≤ = LESS THAN
 MICHOLSON & ASSOCIATES
 606-675 WEST Hastings STREET
 VANCOUVER, B.C.


 ECO-TECH LABORATORIES LTD.
 Frank J. Pessutti, A.Sc.T.
 B.C. Certified Assayer



ECO-TECH LABORATORIES LTD.

ASSAYING - ENVIRONMENTAL TESTING

10041 East Trans Canada Hwy., Kamloops, B.C. V2C 2J3 (604) 573-6700 Fax 573-4667

AUGUST 20, 1991

CERTIFICATE OF ASSAY ETK 91-624

TEUTON RESOURCES CORP.
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1N2

SAMPLE IDENTIFICATION: 38 ROCK samples received AUGUST 12, 1991

----- PROJECT: TWILIGHT

ET#	Description	AG	AG
		(g/t)	(oz/t)
24 -	81524	41.8	1.22


FRANK J. PEZZOTTI
ECO-TECH LABORATORIES LTD.
Certified Assayer

ECO-TECH LABORATORIES LTD.
10041 EAST TRANS CANADA HWY.
RAMLOOPS, B.C. V2C 2J3
PHONE - 604-573-5700
FAX - 604-573-4557

AUGT 20, 1991

TEUTON RESOURCES CORP. - ETK 91-624
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1M2

UES IN PPM UNLESS OTHERWISE REPORTED

PROJECT: TWILIGHT
38 ROCK SAMPLES RECEIVED AUGUST 12, 1991

	DESCRIPTION	AD(ppb)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU FE(%)	E(%)	LA MG(%)	MM	MO NA(%)	NI	P	PB	SB	SM	SR TI(%)	U	V	W	X	ZN
1-	01501	20	3.4 .60	225	6	60	< 5 1.75	7	24	59	631	5.12	.40	<10	.39	1713	3	.09	9	1520	31	<5 <20	42	.01 < 10	25 <10	8 131
2-	01502	75	.6 .35	205	10	60	5 .80	8	12	140	603	3.14	.28	<10	.07	899	10	.04	7	790	22	<5 <20	10 <.01 < 10	14 <10	4 70	
3-	01503	15	4.0 .55	270	6	45	< 5 4.57	8	14	109	486	4.10	.23	<10	.91	1916	7	.05	6	790	11	5 <20	69 <.01 < 10	22 <10	6 54	
4-	01504	10	2.4 .38	275	8	55	< 5 1.30	8	10	61	269	3.58	.24	<10	.23	831	4	.04	3	910	13	5 <20	26 <.01 < 10	17 <10	4 40	
5-	01505	5	3.2 .47	350	6	65	< 5 3.87	9	11	52	285	4.36	.33	<10	.75	1929	2	.05	4	1370	11	5 <20	101 <.01 < 10	20 <10	7 40	
6-	01506	10	4.2 .64	65	4	65	< 5 4.61	2	17	37	685	4.51	.44	<10	1.14	1915	2	.06	6	1880	11	<5 <20	127 .01 < 10	25 <10	9 81	
7-	01507	70	7.2 .43	185	12	55	< 5 2.33	5	9	178	587	2.72	.31	<10	.43	943	11	.03	6	960	8	15 <20	50 <.01 < 10	16 1020	10 32	
8-	01508	20	2.4 .68	130	8	65	< 5 4.11	4	17	31	186	4.47	.46	<10	.86	1503	2	.05	6	1890	12	10 <20	87 .02 < 10	26 580	9 54	
9-	01509	115	6.0 .70	100	8	80	5 3.81	3	24	62	795	4.85	.47	<10	.68	1440	4	.06	7	1770	14	<5 <20	86 .03 < 10	27 <10	9 73	
0-	01510	90	11.4 .68	90	12	60	< 5 3.64	4	24	65	2559	5.17	.49	<10	.84	1345	4	.06	8	2450	17	<5 <20	124 .02 < 10	30 <10	8 142	
1-	01511	75	13.0 .61	145	8	50	5 3.92	4	22	59	2151	5.12	.39	<10	.79	1707	6	.08	4	2140	9	5 <20	115 .01 < 10	26 10	8 114	
2-	01512	35	9.4 .53	85	10	50	< 5 5.10	3	19	50	695	4.97	.36	<10	1.13	1886	4	.06	6	1840	8	5 <20	148 .01 < 10	19 170	8 77	
3-	01513	30	3.0 .76	140	10	75	< 5 5.48	4	19	43	596	5.03	.53	<10	1.10	1918	3	.06	7	2210	9	<5 <20	212 .02 < 10	27 10	10 75	
4-	01514	25	4.6 1.06	60	6	75	5 4.90	2	22	41	1485	5.14	.58	<10	1.16	1489	2	.07	6	2060	6	<5 <20	128 .05 < 10	39 <10	10 94	
5-	01515	10	4.0 .96	75	10	85	< 5 3.81	3	24	54	1601	5.17	.60	<10	.83	1417	4	.06	7	2070	7	5 <20	93 .04 < 10	38 20	8 77	
6-	01516	100	2.6 .71	30	10	80	< 5 4.30	1	18	57	653	4.84	.47	<10	.86	1954	6	.06	7	1860	12	<5 <20	88 .02 < 10	29 10	9 58	
7-	01517	5	1.8 .99	35	8	65	< 5 3.40	1	23	28	746	5.13	.43	<10	1.08	1523	2	.08	7	1990	9	5 <20	76 .03 < 10	36 <10	8 75	
8-	01518	5	1.2 .84	175	12	95	< 5 3.07	5	18	40	471	4.61	.53	<10	.63	1630	3	.06	6	2040	7	<5 <20	86 .02 < 10	30 <10	8 66	
9-	01519	10	2.4 1.39	10	8	45	< 5 4.12	1	50	25	276	5.10	.70	<10	.85	1917	2	.11	12	1920	64	10 <20	97 .07 < 10	51 <10	9 92	
0-	01520	5	.6 2.70	15	10	145	< 5 6.13	1	25	21	211	4.70	.57	<10	1.53	1756	2	.06	10	1870	8	10 <20	123 .08 < 10	78 <10	13 121	
1-	01521	5	1.2 1.94	25	12	105	< 5 6.74	1	25	36	439	4.43	.71	<10	1.75	1749	5	.07	10	1530	9	10 <20	152 .10 < 10	81 <10	12 116	
2-	01522	15	2.6 1.74	30	12	85	< 5 3.24	1	21	79	618	3.80	.01	<10	1.03	1247	5	<.01	8	1750	11	5 <20	63 .02 < 10	60 <10	1 82	
3-	01523	5	2.0 1.67	20	8	115	5 3.73	1	20	40	604	3.70	.66	<10	1.04	1509	3	.06	7	1600	10	5 <20	63 .08 < 10	59 <10	9 84	
4-	01524	110	>30. .06	10	10	5	20 .02	7	11	160	6756	2.88	.02	<10	.03	45	9	.03	2	5040	39	<5 <20	2 <.01 < 10	6 <10	<1 176	
5-	01525	10	1.4 2.48	10	12	125	< 5 3.99	1	24	42	614	3.70	1.24	<10	1.75	1140	2	.08	13	1810	7	5 <20	160 .19 < 10	134 <10	9 79	
6-	01526	10	1.4 2.97	15	8	110	< 5 4.18	1	26	36	402	3.70	.60	<10	2.05	1180	<1	.09	13	1740	6	5 <20	214 .10 < 10	145 <10	9 101	

TON RESOURCES CORP.

RE 2

ECO-TECH LABORATORIES LTD.

AUGUST 20, 1991

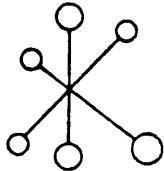
	DESCRIPTION	AU(ppb)	AG AL(%)	AS	B	BA	BI CA(%)	CD	CO	CR	CU FE(%)	E(%)	LA MG(%)	HM	HO NA(%)	MI	P	PB	SB	SM	SR TI(%)	U	V	W	Y	ZN				
17-	81527	5	.8	2.79	10	6	95	< 5	6.16	<1	28	40	354	3.70	.52	<10	1.87	1367	2	.09	17	1800	4	5	<20	341	.10 < 10	171 < 10	11	88
18-	81528	40	1.6	2.73	10	10	95	< 5	5.04	1	27	43	919	3.70	.44	<10	1.88	1245	2	.08	14	1920	6	5	<20	228	.09 < 10	156 < 10	11	106
19-	81529	5	.4	2.61	10	10	70	< 5	5.61	1	28	30	229	3.93	.34	<10	1.87	1326	2	.08	13	2000	6	5	<20	306	.07 < 10	146 < 10	12	73
20-	81530	15	1.2	2.90	15	10	120	< 5	4.05	1	29	46	521	3.72	.69	<10	2.17	1253	3	.08	14	1800	9	10	<20	183	.12 < 10	127 < 10	9	105
21-	81531	5	.6	3.71	10	8	115	< 5	4.66	1	34	32	240	3.67	.79	<10	2.84	1420	1	.10	18	2280	5	10	<20	174	.14 < 10	177 < 10	12	112
22-	81532	5	.6	3.30	10	6	105	< 5	4.09	1	32	59	338	3.68	.51	<10	2.72	1097	4	.10	14	2040	6	10	<20	198	.10 < 10	201 < 10	11	104
23-	81533	35	6.8	2.66	15	10	90	< 5	4.77	2	26	34	1326	3.92	.58	<10	2.01	1177	1	.09	14	2030	6	5	<20	271	.10 < 10	153 < 10	11	136
24-	81534	5	1.0	1.89	20	10	75	< 5	5.50	1	25	47	557	3.70	.96	<10	1.12	1498	3	.07	7	2040	5	5	<20	167	.13 < 10	71 < 10	11	63
25-	81535	5	1.0	2.94	15	16	275	< 5	3.67	1	32	31	369	3.69	1.60	<10	2.44	1286	3	.09	11	2090	5	5	<20	181	.28 < 10	221 < 10	10	112
26-	81536	5	.6	2.29	10	14	150	< 5	6.27	1	27	32	405	3.70	1.04	<10	2.13	1496	1	.07	12	1990	5	10	<20	190	.17 < 10	114 < 10	11	81
27-	81537	5	.4	2.21	15	16	250	< 5	6.59	1	26	74	230	3.70	1.05	<10	2.14	1493	5	.06	10	2300	4	5	<20	210	.16 < 10	128 < 10	12	83
28-	81538	10	.8	2.21	25	8	95	< 5	5.12	1	26	64	421	3.94	.76	<10	1.73	1494	3	.07	13	2290	7	5	<20	195	.14 < 10	124 < 10	11	99

#: < = LESS THAN

NICHOLSON & ASSOCIATES
606-675 WEST HASTINGS STREET
VANCOUVER, B.C.



ECO-TECH LABORATORIES LTD.
Frank J. Perszotti, A.Sc.T.
B.C. Certified Assayer



ECO-TECH LABORATORIES LTD.

ASSAYING - ENVIRONMENTAL TESTING

10041 East Trans Canada Hwy., Kamloops, B.C. V2C 2J3 (604) 573-5700 Fax 573-4557

OCTOBER 18, 1991

CERTIFICATE OF ASSAY ETK 91-816

TEUTON RESOURCES CORP.
602 - 675 WEST HASTINGS STREET
VANCOUVER, B.C.
V6B 1N2

SAMPLE IDENTIFICATION: 57 ROCK samples received OCTOBER 8 , 1991

-----PROJECT: DEL NORTE

ET#	Description	AU (g/t)	AU (oz/t)	AG (g/t)	AG (oz/t)	CU (%)	ZN (%)	PB (%)	AS (%)
1-	ERK 250	3.09	.090	-	-	-	-	-	-
2-	ERK 251	7.68*	.224	32.4	.95	1.22	-	-	-
3-	ERK 252	4.36	.127	-	-	-	-	-	-
4-	ERK 253	4.18	.122	-	-	-	-	-	-
5-	ERK 254	1.01	.029	-	-	-	-	-	-
6-	ERK 255	4.77	.139	-	-	-	-	-	-
13-	ERK 262	-	-	-	-	-	2.20	-	-
23-	ERK 272	-	-	-	-	-	5.44	1.22	-
30-	ERK 280	7.68*	.224	-	-	-	-	-	6.71
32-	ERK 282	8.68*	.253	-	-	-	-	-	4.73
33-	ERK 283	6.52*	.190	-	-	-	-	-	2.05
34-	ERK 284	3.59	.105	-	-	-	-	-	-
35-	ERK 285	10.78*	.314	-	-	-	-	-	2.42
36-	ERK 286	9.04*	.264	219.4	6.40	-	1.38	-	-
51-	VZ 01	-	-	-	-	-	3.08	2.76	-
53-	VZ 02	-	-	-	-	-	-	1.06	-
55-	VZ 03	-	-	31.4	.92	-	5.46	5.86	-

NOTE: * = SAMPLE SCREENED AND METALLIC ASSAYED


ECO-TECH LABORATORIES LTD.
Frank J. PEZZOTTI, A.Sc.T.
B.C. Certified Assayer

METALLIC CALCULATION

SAMPLE NUMBER	-140 VALUE	+140 VALUE	CALCULATED VALUE
816-2	7.53	10.4275	7.684123
816-4	4.67	3.240614	4.18391
816-6	5	4.251869	4.779802
816-30	7.43	13.31516	7.687914
816-32	9.51	6.076854	8.684569
816-33	7.13	5.26692	6.528433
816-35	10.95	8.289474	10.78083
816-36	7.42	22.57317	9.041694

APPENDIX IV
SAMPLE DESCRIPTIONS

ROCK SAMPLE DESCRIPTION RECORD

DEL NORTE PROPERTY

- 81501-18 1.0 m continuous chips.
- 81519-23 1.0 m continuous chips.
- 81751 Mottled crystal tuff, tan weathered, 2 - 3% disseminated pyrite; 1% disseminated galena, hematite; minor chlorite altered.
- 81752 Quartz/calcite vein in crystal tuff, 1 - 2% disseminated galena; 1 - 2% disseminated sphalerite, trace chalcopyrite; 3 - 4% disseminated pyrite.
- 81753 1 m chip; shear zone in tuff, oxidized with arsenic staining.
- 81754 1 m chip - same as 81753.
- 81755 1 m chip; same shear, more intense fracture; trace arsenopyrite.
- 81756 Float; milky quartz vein, 30 cm wide; vuggy with carbonate altered; trace galena.
- 81757 1 m chip; shear zone, 3 m wide, vuggy, oxidized, quartz veined.
- 81758 Same 1 m chip.
- 81759 Same 1 m chip.
- 81760 Networked quartz/carbonate veins in tuff; trace pyrite, sphalerite in veins, host tuff has 2% disseminated pyrite.
- 81761 Shear zone in tuff; limonitic altered with 2% disseminated pyrite, trace arsenopyrite.
- 81762 Shear zone with pods, ~ 1 m x 40 cm of massive pyrite - trace arsenopyrite, chalcopyrite.
- 81763 Same shear, siliceous with 2% disseminated pyrite, chalcopyrite, limonitic altered - 1 m chip.
- 81764 Same shear, 10 m upsection - 1 m chip.
- 81765 Same 1 m chip.

- 81766 Same 1 m chip.
- 81767 Shear zone, with two small (9 cm x 15 cm) veins, highly oxidized with 5 - 10% chalcopyrite, malachite; 1 m chip.
- 81768 Same 1 m chip.
- 81769 Same, grab sample.
- 81770 9 m upsection, splays into a zone of drusy, crystallized quartz/carbonate veining; 2 - 4% pyrite, chalcopyrite on vein edges, limonitic staining - 1 m chip.
- 81771 Same, grab sample.
- 81772 Highly oxidized, limonitic stained tuff with pods (10cm x 15cm) of 4 - 5% pyrite, trace chalcopyrite, arsenopyrite.
- 81773 Sheared, kaolinized porphyry with small (40cm x 10cm) raft of tuff, limonitic altered.
- 81774 Small (20cm x 2m?) quartz/carbonate vein, slightly vuggy, heavy limonitic altered; trace galena, sphalerite.
- 81775 Shear/fault in tuff, limonitic altered with 2% disseminated pyrite.
- 81776 Flat - quartz/carbonate vein, thickness unknown, bladed carbonate crystals; 2 - 5% massive galena on vein edges.
- 81777 Quartz/carbonate vein in dark green ash tuff, slightly vuggy (1%) - good crystal growth.
- 81778 Float - massive milky quartz vein, may be insitu; limonitic, vuggy and fractured.
- 81779 Shear zone in crystal tuff, mottled green-black; representative sample.
- 81780 Networked quartz/carbonate veins in mottled tuff, next to contact, limonitic and vuggy - 2.5 cm wide.
- 81781 1 m chip; contact between argillite/tuff; networked quartz veins; first 50 cm arsenic stain in argillite/siltstone.
- 81782 1 m chip, same.
- 81783 1 m chip, same.
- 81784 Grab, volc. side of contact.

- 82522 Float boulder; 30 cm x 15 cm angular intensely quartz-sericite altered, weakly kaolinized; rusty orange-yellow on weathered surface; 5 - 10% disseminated fine to medium grained pyrite.
- 82523 Float boulder; 10 cm x 5 cm angular milky white quartz vein; discrete banding (0.2 - 1.0 cm wide) of fine to medium grained pyrite +(-) trace galena +(-) trace arsenopyrite.
- 82524 Random grab; malachite stain along phyllitic planes in phyllitically altered crystal tuff; 1 - 3% disseminated fine grained pyrite.
- 82525 Select grab; 5 cm wide erratic quartz-carbone; sulphide fracture; fracture at 030/50 SW; contains medium to coarse grained pyrite, chalcopyrite (10 - 20%), malachite stain.
- 82526 30 cm chp; quartz-carbonate-sulphide fracture at 335/85 NE; fine to medium grained pyrite (5 - 10%); trace - 2% chalcopyrite; trace - 3% arsenopyrite.
- 82527 Select grab ; quartz- sulphide pod, milky white quartz with disseminated medium grained pyrite and chalcopyrite (5 - 10%) +(-) fine grained arsenopyrite, minor malachite stain.
- 82528 1.10 m chip; fracture zone at 040 with several 1 - 2 cm wide quartz-sulphide stringers; disseminated pyrite and chalcopyrite within quartz stringers.
- 82529 40 cm chip; quartz-sulphide fracture running at 020/40 NW; quartz and medium to coarse grained pyrite and chalcopyrite (5 - 15%) +(-) fine grained arsenopyrite oxidized yellow-red on weathered surface.
- 82530 Select grab; narrow mineralized fractures (erratic) within weakly propylitized crystal tuff; < 1 cm wide quartz stringers with disseminations and blebs of fine to medium grained pyrite (2 - 3%) and chalcopyrite (trace - 2%).
- 82531 Select grab; erratic quartz-sulphide fracture ~ 5 - 10 cm wide; disseminated to semi-massive fine to coarse grained pyrite, blebs and inclusions of fine grained chalcopyrite +(-) arsenopyrite.
- 82532 Select grab; erratic quartz-sulphide fracture ~ 5 cm wide; approx. 5 m NW of 82531 (on strike) as per 82531.
- 82533 Select grab; quartz-sulphide fracture in weakly

- propylitized tuff; fracture at ~ 025/60 SW ~ 5 cm wide; vuggy, jarostically stained quartz; blebs of medium grained chalcopyrite; disseminations and blebs of fine to medium grained pyrite; some malachite stain on weathered surface.
- 82534 Select grab of mineralized shear zone of 82535.
- 82535 75 cm chip; quartz-chalcopyrite-pyrite shear running at 332/vertical; quartz-jarosite with abundant fine grained disseminated pyrite and fine to medium grained blebs of chalcopyrite.
- 82536 Select grab; 5 cm wide quartz vein at 310/48 SW. Quartz vein containing bands and fragments of feldspathized country rock (tuff); medium to coarse grained blebs of arsenopyrite (5 - 10%); disseminated fine to medium grained pyrite (1 - 5%); blebs of fine to medium grained chalcopyrite as intergrowths within asp (trace - 2%).
- 82537-45 1.0 m continuous chips.
- 82546-62 1.0 m continuous chips.
- ERK-250 Quartz stringers with massive pyrite and sphalerite; blebby chalcopyrite and some galena; grab from trench.
- ERK-251 As above, high grade; strong malachite stain, much more quartz.
- ERK-252 1 m trench chips; broken, weathered, green altered, with stringers and dissemination of chalcopyrite; abundant specularite.
- ERK-253 1 m - same as ERK-252.
- ERK-254 1 m - same as ERK-252.
- ERK-255 1 m; minor quartz stringers with coarse chalcopyrite stringer; minor specularite.
- ERK-256 1m: altered rock with minor chalcopyrite; fault gouge on west end.
- ERK-257 1m; one small chalcopyrite stringer at West end; rest of rock is green carbonate altered rock with pyrite and malachite stain.
- ERK-258 1m; same as ERK 257
- ERK-277 (Trench--SE of Crackle Zone); 2 m chip; pyrite, chalcopyrite, pyrrhotite, arsenopyrite, quartz and

sulphides.

- ERK-279 3 m chip; 3 - 4 quartz sulphide stringers in crystal tuff.
- ERK-280 Grab of 6" quartz stringer with coarse arsenopyrite and pyrrhotite and chalcopyrite on north end of trench.
- ERK-281 Trench; stringers of quartz with arsenopyrite, chalcopyrite, pyrrhotite, pyrite in 1 m zone.
- ERK-282 Grab of quartz stringer approx. 8 cm with coarse arsenopyrite and chalcopyrite and zinc sulphide; sulphides approx. 20%.
- ERK-283 Grab of quartz with coarse arsenopyrite stringer, approx. 18 cm in zone; up to 1 m wide.
- ERK-284 Trench; 3 m chip; quartz stringers up to 15 cm with coarse pyrrhotite, arsenopyrite with minor chalcopyrite, zinc sulphide and pyrite.
- ERK-285 Grab of 15 cm wide quartz vein with approx. 30% pyrrhotite, arsenopyrite, zinc sulphide, chalcopyrite and pyrite.
- ERK-286 1 m chip at DNSR 039; 3 m south of 117. Quartz stringers with massive chalcopyrite and specularite stringer; chalcopyrite approx. 5%; manganese stained.
- ERK-287-
294 Each 1 m; sheared, manganese stained feld porphyry; abundant pyrrhotite, pyrite; minor chalcopyrite and malachite.

- 81785 Grab, sed. side of contact.
- 81786 W side of contact, networked quartz veins (15-20% with sheared tuff, pods (3-5% of galena/sphalerite, conc. in vein selvages - 1 m chip).
- 81787 Same, grab.
- 81788 8 m down vein from 81786 just before contact - 1 m chip.
- 81789 Sed. side of contact; black, limonitic and fractured; 1 - 5% quartz/carbonate stringers, 1 - 2 m wide.
- 81790 Large shear zone at contact; 20 - 30% quartz stringers/veinlets; trace chalcopyrite, 1% sphalerite, hydro-zincite? on fractured surfaces, in siltstone/argillite; 1 m chip.
- 81791 8 m upslope, large (>50 cm) quartz vein at contact, limonitic altered in vugs (5%).
- 81792 Grab, same.
- 81793 Grab, same.
- 81794 Light green, mottled tuff/porphyry at contact, carbonate altered; 1 m chip.
- 81795 Small shear (30 - 60cm), limonitic with trace galena, sphalerite; 1 m chip.
- 81796 Orange weathering tuff, bleached and feldspathized pods (2 - 5 cm) of oxidized pyrite? up to 2%; 1 m chip.
- 81803 Small shear/veinlet in mottled green tuff with 5 - 7% pyrite; trace - 2% arsenopyrite. Good crystal stal. development in vugs.
- 81804 Quartz breccia - 65 cm x 65 m; clasts 1 - 5 cm, volcanized with fractured surfaces slightly oxidized; trace arsenopyrite (?); grab sample.
- 81805 Arsenic/limonite stained zone in siliceous crystal tuff, vuggy (5 - 10%) with 5% pyrite; grab sample.
- 81806 Brown carbonate altered with pods; disseminated pyrite (5%); chalcopyrite(?) 1%; vuggy; grab sample.
- 81807 Pyritized quartz vein/breccia in tuff, 5 - 10% pyrite in 2 cm pods/disseminated; trace chalcopyrite; grab sample.
- 81808 Oxidized quartz/carbonate vein; <10% pyrite in 1 cm

- veinlets; slightly vuggy.
- 81809 Float - quartz breccia vein, malachite stained, 5 - 10% chalcopyrite disseminated and in pods; slightly vuggy (5%).
- 81810 Representative sample of mottled green black tuff, quartz/carbonate altered with 5% pyrite disseminated and blebs.
- 81811 Representative sample, calcite veined black crystal tuff, slight shear and faulted (?).
- 81812 Mottled green siliceous tuff, 5 - 10% calcite infilling, trace chalcopyrite along microfractures and disseminated.
- 81813 Quartz vein, 5 - 15 cm with 5% chalcopyrite, pyrite, trace arsenopyrite, vuggy; oxidized; grab sample.
- 81814-17 1 m continuous chips (see trench plans).
- 81818-20 1 m continuous chips (see trench plans).
- 81821-37 1 m continuous chips (see trench plans).
- 81838 Float - slightly siliceous green tuff - limonitic altered, 2 - 5% massive disseminated galena; 7 - 10% honey-brown sphalerite; trace chalcopyrite.
- 81839 Outcrop of 81838, vein-like trend, maroon-green intermediate volcanized intense siliceous - 5 - 7% sphalerite and hydrozincite; 2 - 3% galena; trace chalcopyrite; slightly vuggy and irregular on weathered; no obvious contact features; 1 m chip.
- 81840 Same, 1 m chip.
- 81841 Same outcrop as 81839; 1 m chip.
- 81842 Chalcopyrite/malachite stained zone on edge of 10 - 15 m milky quartz vein; 5 - 7% chalcopyrite, maximum 80cm x 50 cm; grab sample.
- 81843-48 Trench #3; 1 m continuous chip.
- 82715 Representative sample of limonitic zone in lapilli tuff; 5% pyrite; 1 - 2 cm cross-cutting quartz/carbonate veins; fractured and recessive, brown soil.
- 82716 Brecciated quartz vein, 30 - 50 cm wide. slightly vuggy and limonitic; trace pyrite, sphalerite (?); 1 m chip.

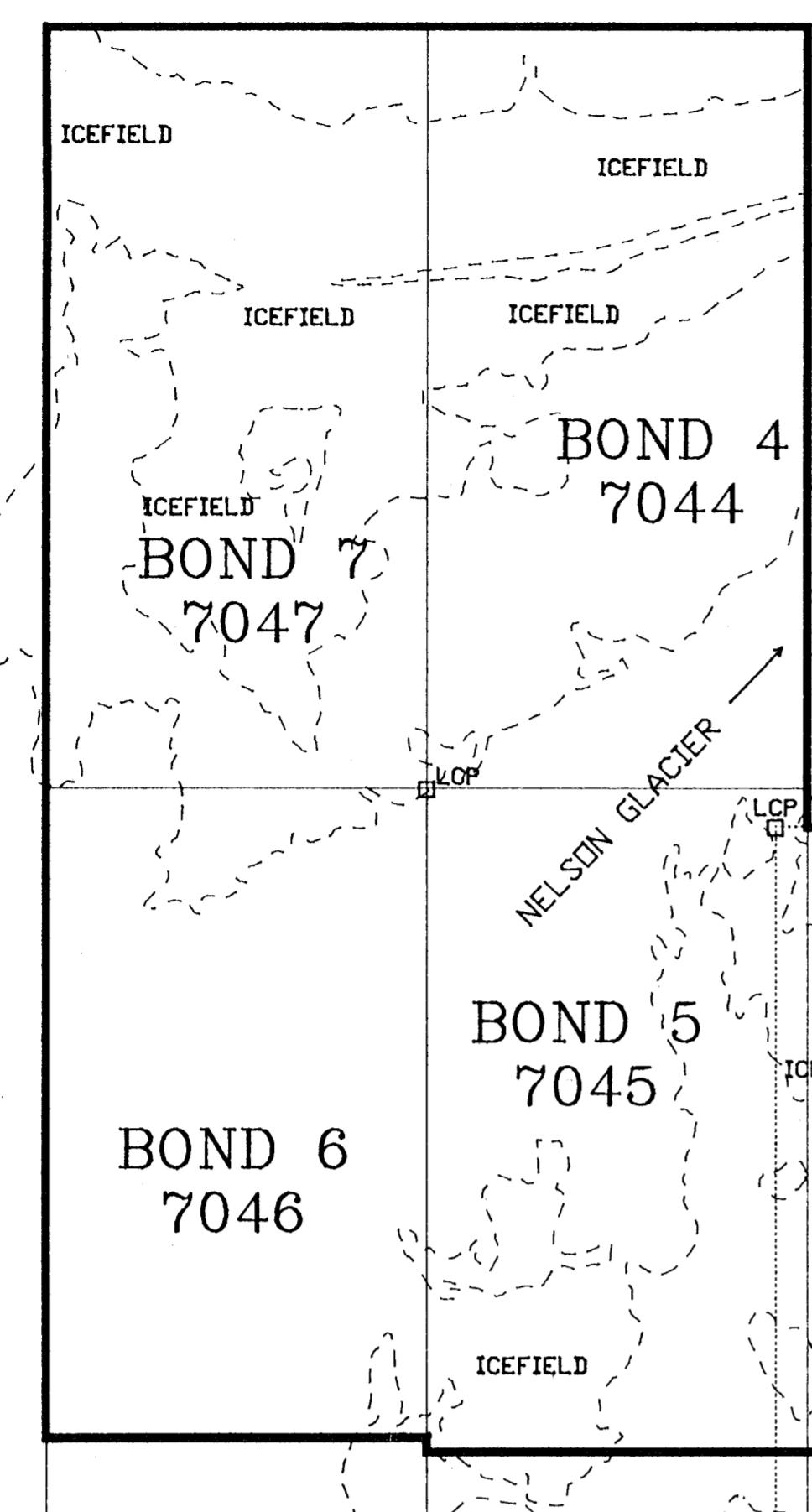
- 82717 Quartz/carbonate vein, recessive, limonitic altered; trace pyrite, sp.hem.; approx. 10 - 20 cm thick; representative sample.
- 82601-18 1.0 m chips across shear zone on (footwall?) felsite dyke; sheared, gougey felsite, with pyrite boxwork, limonitic sections, approx. 4% pyrite on fresher sections.
- 82619 1.0 m contact with fresh unaltered dyke.
- 82620 HW contact of same dyke, sheared crystal tuff with trace lead oxide and approx. 4% pyrite. Grab.
- 82621 1.5 m chip, sheared tuff; Ads. stain pyrite approx. 5%
- 82622 1.5 m chip, contiguous with 82621; sheared tuff.
- 82623-27 1.0 m chip.
- 82628 Grab sample, quartz stringer with traces malachite.
- 82629 1.0 m chip, siliceous altered tuff.
- 81630 1.5 m chip; quartz veinlets, oxidized pyrite boxwork.
- 82631 Select grab; siliceous altered tuff with lead sulphide 2%; chalcopyrite blebs approx. 5%; pyrite.
- 82632 Grab; quartz veinlets in tuff, trace lead sulphide, approx. 10% pyrite.
- 82633 Grab, local float, quartz carbonate altered tuff; lead sulphide 2%; trace pyrite.
- 82634 Grab, disseminated pyrite in moderately carbonate altered tuff.
- 82635 Same as 82634.
- 82636 Grab; quartz veinlets, stringers in shear zone.
- 82637 1.0 m chip; possible strike extension of Bullion Zone; quartz carbonate veining in carbonate altered tuff; trace PbS, pyrite and arsenic approx. 2%.
- 82638 Grab; sheared brecciated argillite; quartz carbonate fractures.
- 82639 1.0 m chip; shear zone; quartz carbonate fractures, gougey (10 cm of #306, 1990).

- 82640 Contiguous with 8222639.
- 82641 1.0 m chip; shear zone at contact Salmon Betty; quartz carbonate brecciation.
- 82642 1.0 m chip consisting of 2 shears, some quartz carbonate veining (in argillite).
- 82643 Contiguous with 82642; 1.0 m chip; stronger quartz.
- 82644 Contiguous with 82643; 1.5 m chip.
- 82645 Sheared gougey argillite with some strong pyrite boxwork; 1.0 m chip.
- 82646 Strike extension of NMG vein down slope; 1.0 m chip; gougey, bleached; approx. 5% lead sulphide.
- 82647 Grab; siliceous altered tuff; trace chalcopyrite; lead oxide <1%; pyrite 3%.
- 82648-50 3 at 1.0 m chips across quartz carbonate veined and altered tuff; as stain lead oxide approx. 1%, trace chalcopyrite.
- 82651 1.0 m chip, siliceous shear zone.
- 82652-54 3 - 1.0 m chips, extension of structure in 82651, sheared quartz carbonate altered tuff, strong iron oxide.
- 82660-87 All 1.0 m chips.
- 82689 1.0 m chip; gougey shear zone with quartz fragments.
- 82690 1.0 m chip; shear zone, quartz veining; brecciated argillite.
- 82691-98 1.0 m chip.
- 82699 1.5 m chip.
- 82401 3.0 m shear across shear zone propylatized crystalline tuff.
- 82402 1.5 m continuous chip propylatized crystalline tuff weakly oxidized, trace pyrite, disseminated.
- 82403 as per 82402
- 82404 2.5 m continuous chip, quartz carbonate altered crystalline tuff, weakly sericitized, trace - 1% disseminated pyrite throughout.

- 82405 Float boulder; altered silicified crystalline tuff jarositically altered, fine wispy pyrite throughout.
- 82406 Float boulder; brownish-red weathered, jarostically altered crystalline tuff with trace - 2% disseminated pyrite +(-) arsenopyrite throughout.
- 82407 1 m chip; quartz breccia which cuts through phyllitic argillites, weakly chloritized, trace pyrite disseminated throughout.
- 82408 1.5 m chip; silicified crystall tuff, jarositically altered, carbonate infilling along fractures, trace pyrite +(-) chalcopyrite as disseminations.
- 82409 1.5 m chip; rusty reddish-brown weathered, jarostically altered silicified crystall tuff, with fine to medium grained pyrite +(-) chalcopyrite disseminated throughout and along fractures as infillings.
- 82410 1.5 m chip; as per 82409.
- 82411 Float grab; rusty orange-brown weathered, silicified (weakly sericitized) crystal tuff, wispy pyrite and disseminated pyrite throughout.
- 82412 Random grab; rusty orange-brown weathered, silicified (sericitized) crystal tuff with trace amounts of disseminated pyrite.
- 82413 1 m continuous chip; across shear zone contact; blackish graphitic argillite, very soft and vuggy, trace - 1% disseminated pyrite throughout.
- 82414 Talus float; bleached silicified sericitized crystal tuff which contains fine grained disseminated pyrite +(-) magnetite, chalcopyrite.
- 82415 10 cm wide quartz band/stringer which contains trace chalcopyrite as disseminations with malachite staining along fractures.
- 82416 1.5 m continuous chip; shear zone, moderately silicified with abundant quartz stringers, trace pyrite throughout.
- 82417 as per 82416.
- 82418 1.5 m continuous chip; quartz breccia contact between crystal tuff and laminated siltstone, sheared and moderately fractured and broken up, trace pyrite +(-) galena, sphalerite (footwall).

- 82419 High grade grab of NMG Vein; trace - 10% coarse grained galena, trace - 15% coarse grained sphalerite, trace - 5% pyrite +(-) chalcopyrite.
- 82420 1.0 m chip, hanging wall, quartz breccia shear contact between crystal tuff, laminated siltstone argillite, moderately fractured and broken up; trace pyrite +(-) galena sphalerite.
- 82421 1.0 m continuous chip, jarositically altered crystal tuff, fractured, broken up and sheared, abundant quartz stringer, trace pyrite +(-) chalcopyrite.
- 82422 1.0 m continuous chip as per 82421.
- 82423 1.0 m continuous chip as per 82421.
- 82425 Grab sample; brecciated quartz vein with finely disseminated galena, pyrite disseminated throughout, rusty weathered.
- 82428 Talus float; brecciated epithermal veined stringer, orange-brown weathered, fine grained pyrite weakly disseminated throughout.
- 82429 Talus float; brecciated epithermal veined, which contains very fine grained massive pyrite along breccia rims +(-) arsenopyrite.
- 82501 60 cm chip; quartz - carbonate vein on footwall of Bullion Zone, 325/80 NE, trace disseminated pyrite, limonitic fractures.
- 82502 20 cm chip; brecciated quartz vein at 360/35E. Vuggy quartz vein with angular fragments of carbonate altered country rock; 5 - 7% pyrite, malachite stain.
- 82503 1.10 m chip; sheared contact between feldspar porphyry dyke and silicified, carbonate altered crystal tuff. Shearing at 320/vert. sheared, silicified and carbonate altered crystal tuff; stringers and pods of quartz, limonitic staining.
- 82504 30 cm chip; dip slope contact of feldspar porphyry dyke and crystal tuff; contact at 320/40 SW well brecciated quartz vein with angular fragments of carbonate altered country rock; medium to coarse disseminated pyrite, galena +(-) chalcopyrite.
- 82505 Random grab; silicified tuff; weakly feldspathized, narrow stringers of ankerite, trace disseminated pyrite.

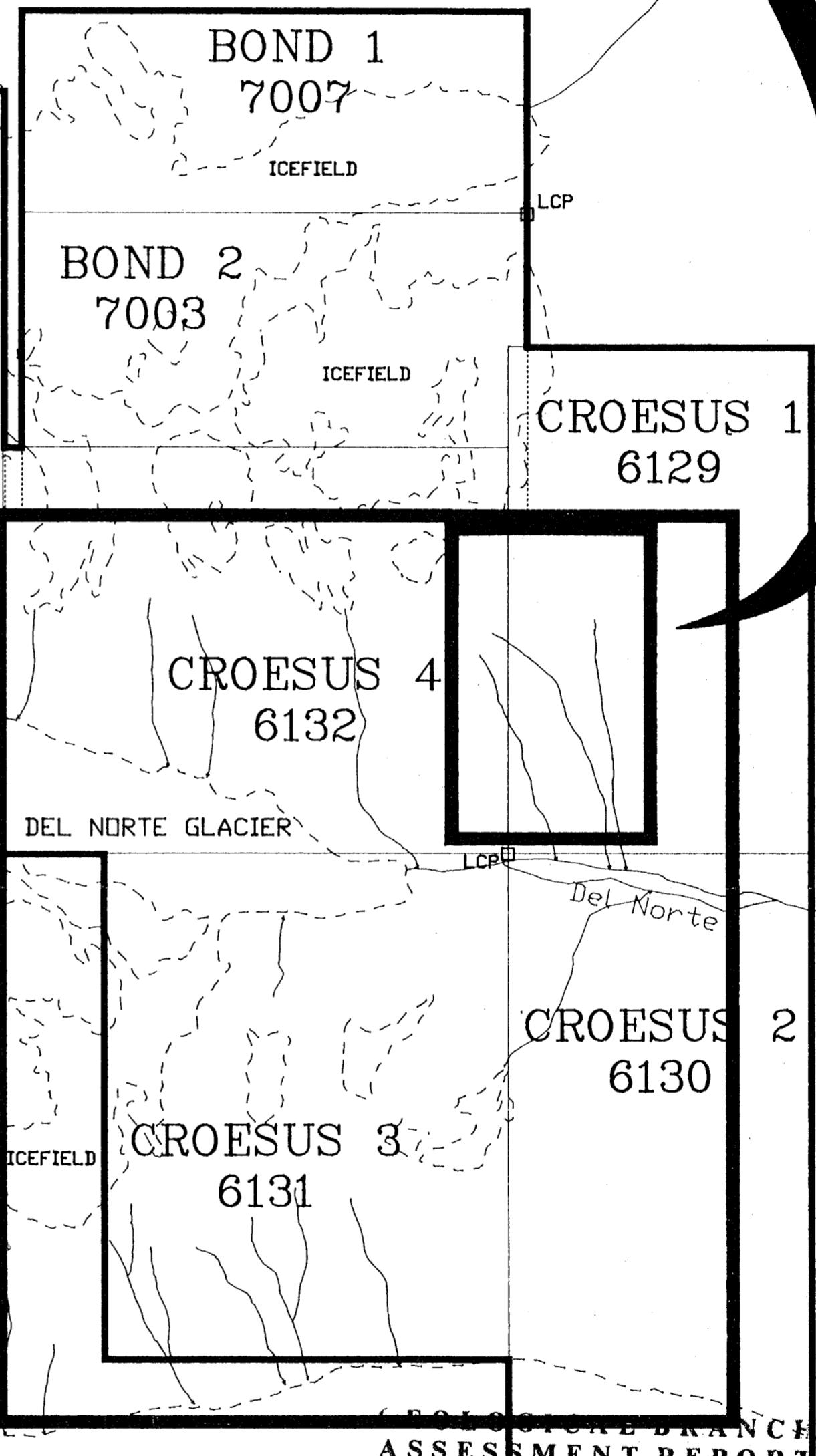
- 82506 Select grab; rusty sheared crystal tuff, weak to moderate yellow-green arsenic staining; 1 - 3% disseminated fine grained pyrite +(-) arsenopyrite.
- 82507 Select grab; 10 cm wide, rusty, mineralized fracture (very erratic) within carbonate altered, weakly silicified crystal tuff. 320/83 NE, fine grained pyrite +(-) arsenopyrite.
- 82508 Random grab; series of erratic shears within feldspar porphyry flow; rusty, narrow fractures (to 10cm) trending 295/72 NE, fine grained pyrite +(-) arsenopyrite.
- 82509 Random grab; rusty, mineralized silty argillite with prevalent yellow-green (arsenic?) stain, occasional narrow quartz-carbonate stringers; 1 - 5% disseminated fine grained pyrite.
- 82510 50 cm chip; rusty mineralized silty argillite, some yellow-green (arsenic?) stain, limonitic stain, 1 - 3% disseminated fine grained pyrite boxwork texture where pyrite leached out. Schistosity at 320/38 NE.
- 82511 Random grab; rusty, mineralized silty argillite on upper contact of feldspar hornblende porphyry dyke (320/85 NE); limonitic and ferro manganese stain; some fine grained disseminated pyrite as stringers and erratic fractures.
- 82512 Select grab; quartz vein approx. 20 cm wide on upper contact of feldspar hornblende porphyry dyke and rusty sediments (340/80 NE); vuggy white quartz, weakly sericite altered, some limonitic alteration along margins; rare disseminated fine grained pyrite.
- 82513 60 cm chip; sheared zone in crystal tuff; shearing at 180/60 W; silicified and feldspathized; 2-5% pyrite +(-) arsenopyrite with arsenic stain; heavily oxidized on surface.
- 82414 1.0 m chip (of 2.0 m continuous chip); shear zone in oxidized, altered crystal tuff; shearing at 110/vert; phyllitic-argillitic, weakly kaolinitized tuff; limonitic stain, some quartz infilling; disseminated fine grained pyrite and rare disseminated galena.
- 82515 1.0 m chip (of 2.0 m continuous chip) as per 82514.
- 82516 75 cm chip; shear zone at 145/80 NE; oxidized phyllitic-argillitic, weakly kaolinitized crystal tuff; vuggy, some quartz-carbonate infilling; 1 - 5% disseminated fine grained pyrite +(-) arsenopyrite; fine disseminations and erratic stringers of galena.



NELSON GLACIER



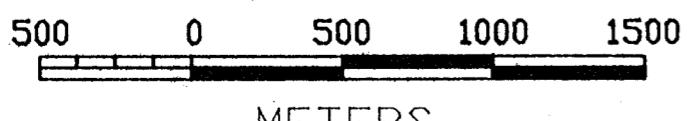
LOCATION OF FIGS. 5-14



GEOLOGICAL BRANCH
ASSESSMENT REPORT

LOCATION OF FIGS. 15-17

WILLOUGHBY GLACIER



Note: All Figures Are In
Skeena Mining Division

22,103

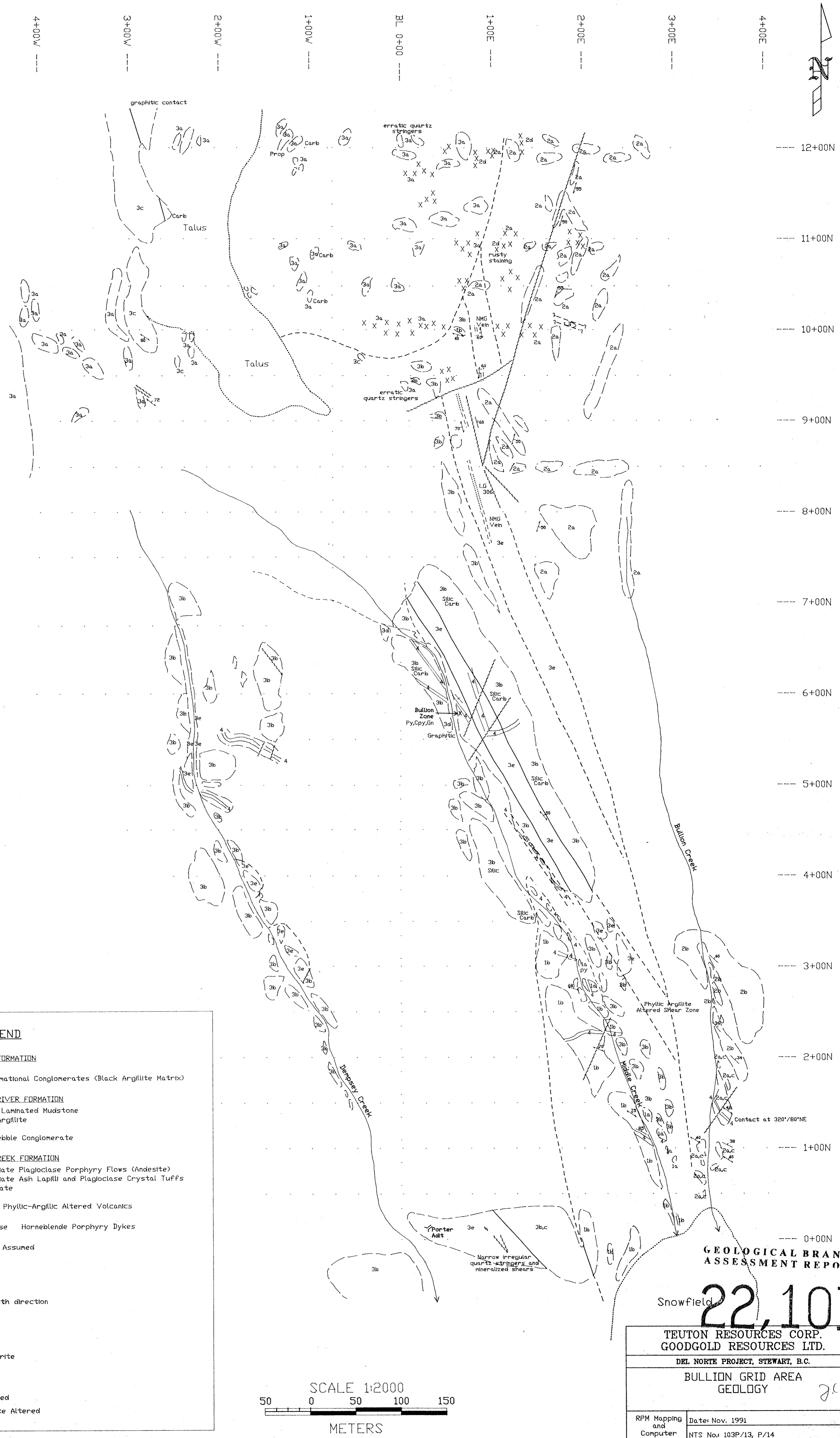
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GOODGOLD RESOURCES LTD.

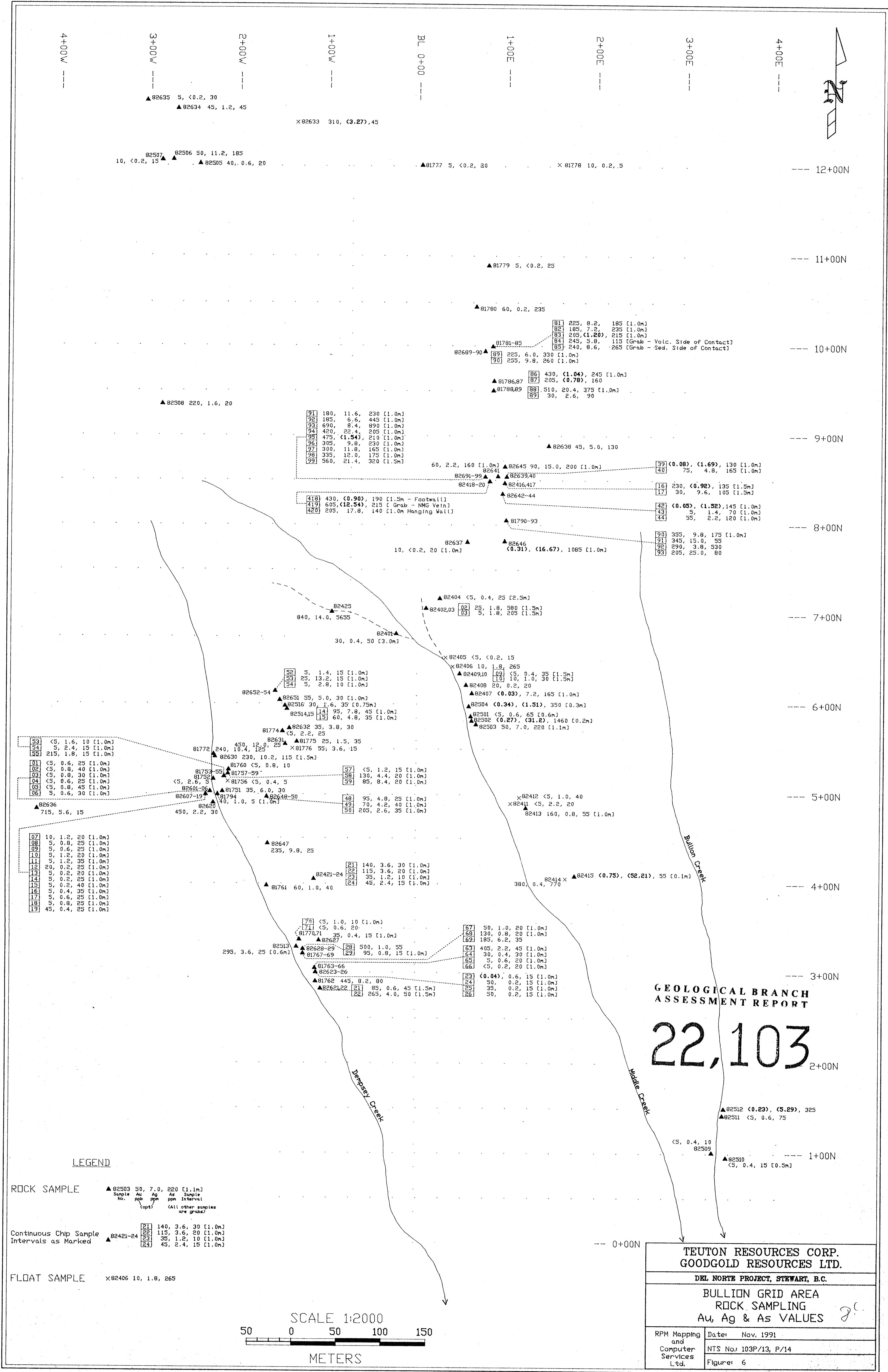
DEL NORTE PROJECT, STEWART, B.C.

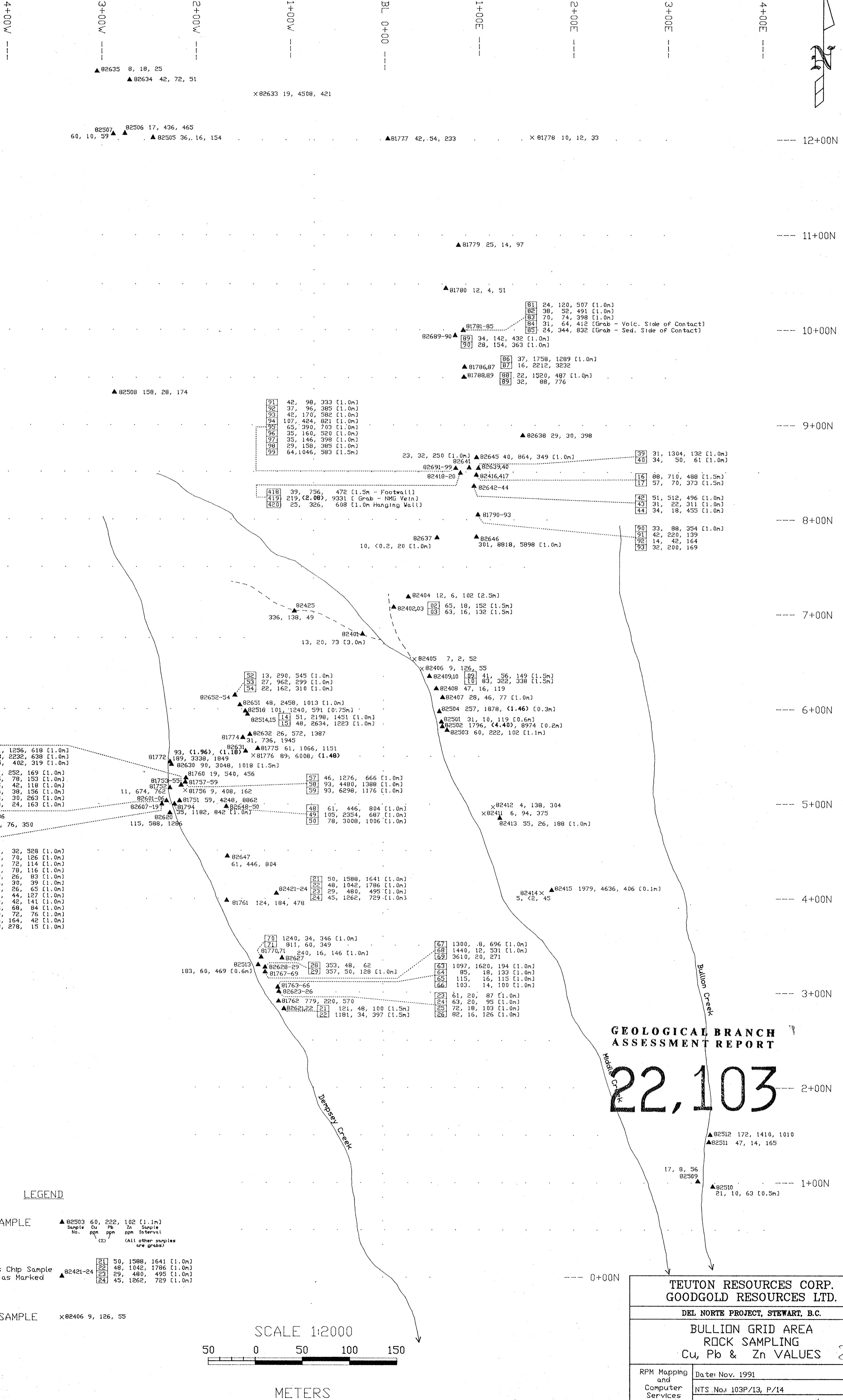
INDEX TO
FIGURES 5-17

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RPM Mapping and Computer Services Ltd.	Date Nov. 1991
NTS No. 103P/13, P/14	
Figure 4	



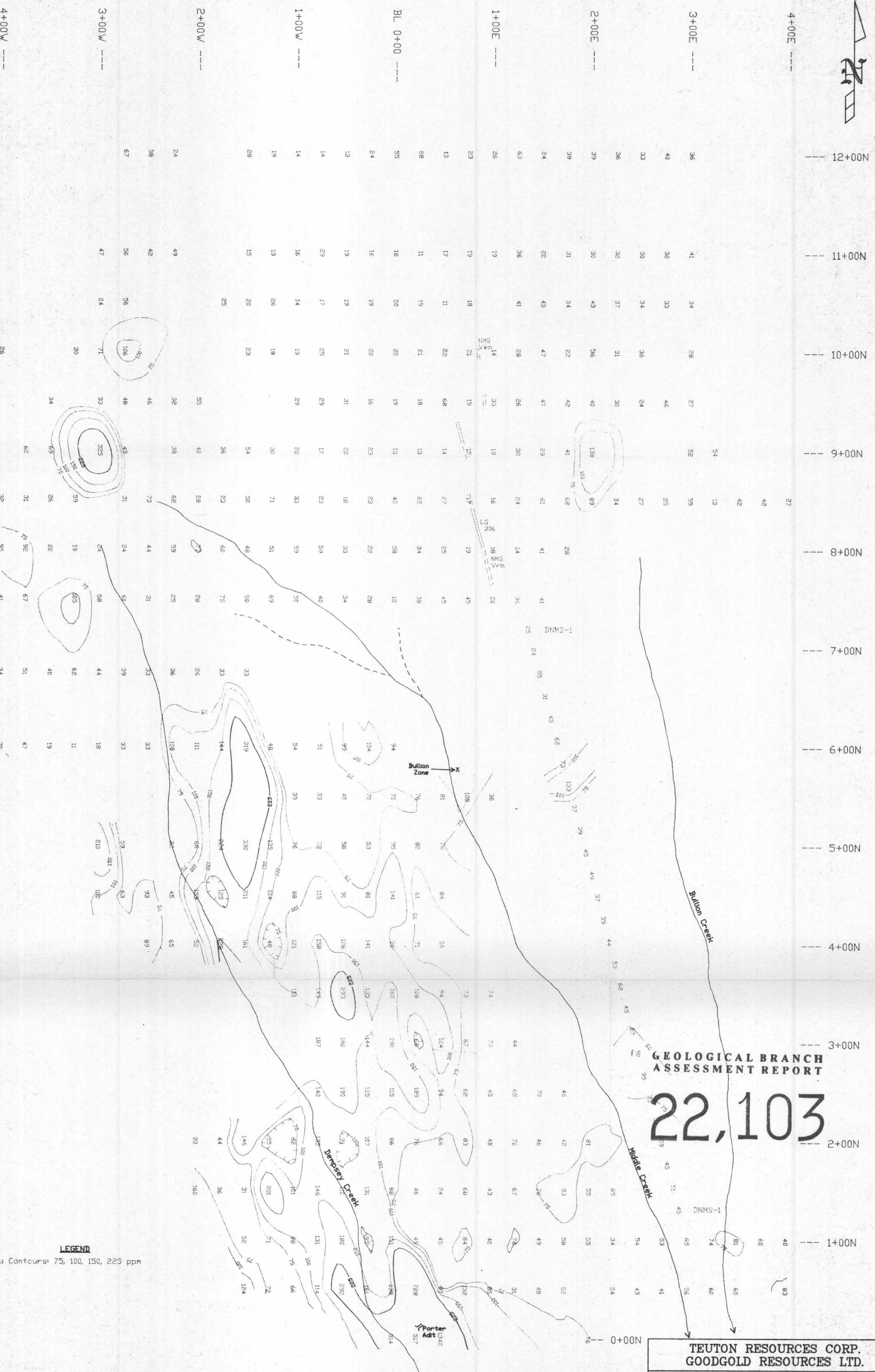


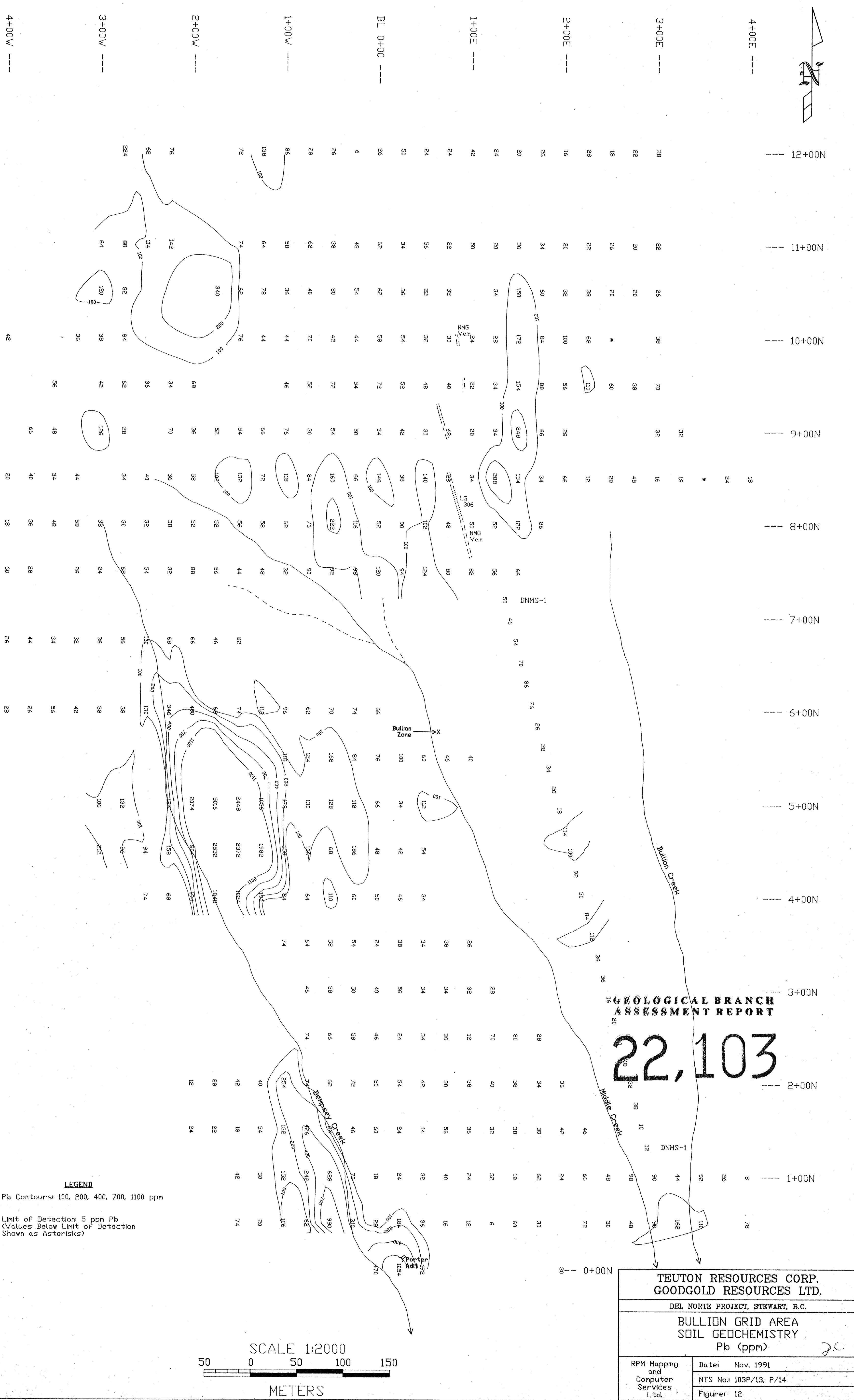


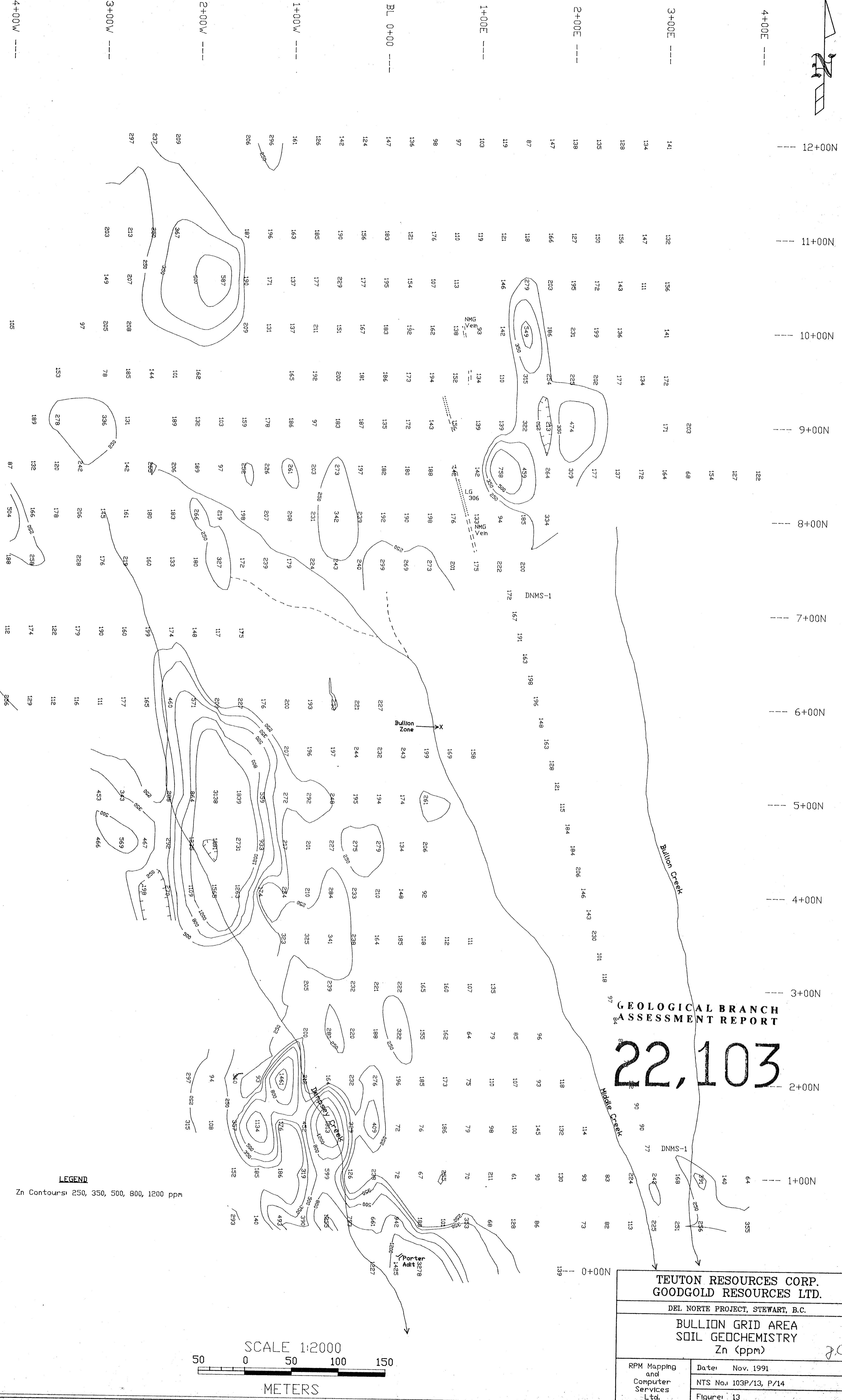


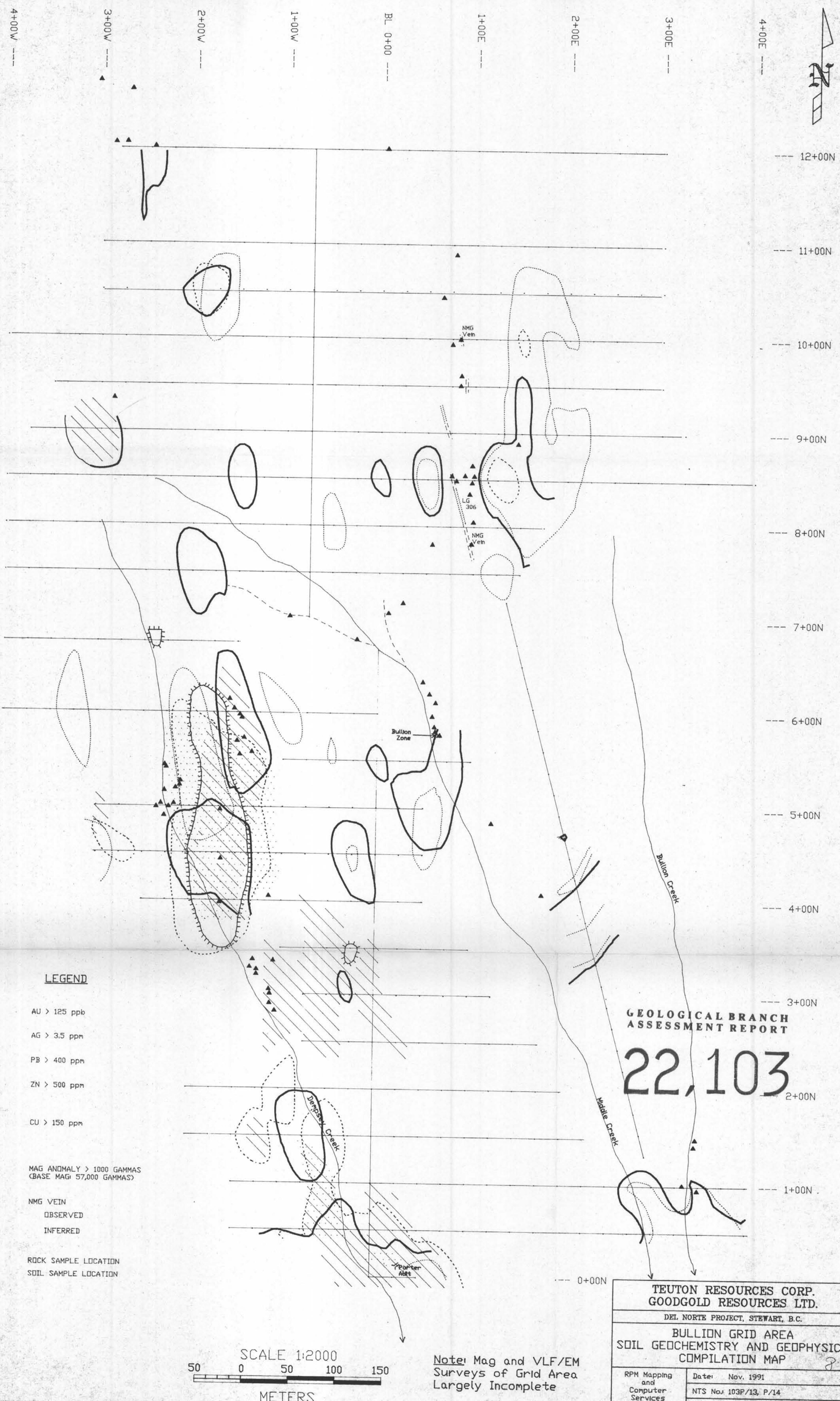












BULLION GRID AREA
See Figs. 6-14

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CRACKLE ZONE
DETAIL MAP
SCALE 1:200

LEGEND

3 Betty Creek Formation
3a Intermediate Plagioclase Porphyry Flow (Andesite)
3b Quartz-Carbonate Stringers Numerous 1-5 cm wide qtz-cpy-py stringers at 020°-060°
3c Agglomerate
3d Argillite
3e Strongly Phyllitic-Agillite Altered Volcanics
3f Sheared Altered Volcanics

Silic Silicified
Feld Feldspatized
Qtz Quartz
Py Pyrite
Cpy Chalcopyrite
Mag Magnetite

40° Orientation of Vein or Shear
Channel Sample with Sample Number

▲ 81813 Rock Sample

Some 1-2 cm wide qtz-cpy-py stringers at 060°-090°

Some 1-2 cm wide qtz-cpy-py stringers at 020°-060°

Some 1-2 cm wide qtz-cpy-py stringers at 060°-090°

DEL NORTE GLACIER

ICEFIELD

