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1989 GEOLOGICAL AND GEOCHEMICAL REPORT
ON THE
SHAKE 1-4 CLAIMS

FILMED

Located in the Telegraph Creek Area
Liard Mining Division
NTS 104G/13E
57° 49' North Latitude
131° 36' West Longitude

-prepared for-
CANDELA RESOURCES LTD.

-prepared by-
David A. Caulfield, F.G.A.C.

September, 1989

19,127

GEOLOGICAL BRANCH
ASSESSMENT REPORT

1989 GEOLOGICAL AND GEOCHEMICAL REPORT ON THE SHAKE 1-4 CLAIMS

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1.0 INTRODUCTION

The Shake 1-4 claims were staked in 1988 to cover favorable geology and gossans on Rugged Mountain, approximately 28 kilometers southwest of Telegraph Creek in northwestern British Columbia (Figure 1). The geological similarity to the Galore Creek, Iskut River, Sulphurets and Stewart mining camps to the south and the area's potential for precious metal mineralization have sparked renewed exploration interest throughout the area.

Reconnaissance exploration, consisting of geological mapping, prospecting and geochemical sampling, was carried out over the Shake 1-4 property during June of 1989. Equity Engineering Ltd. conducted this program for Candela Resources Ltd. and has been retained to report on the results of the fieldwork.

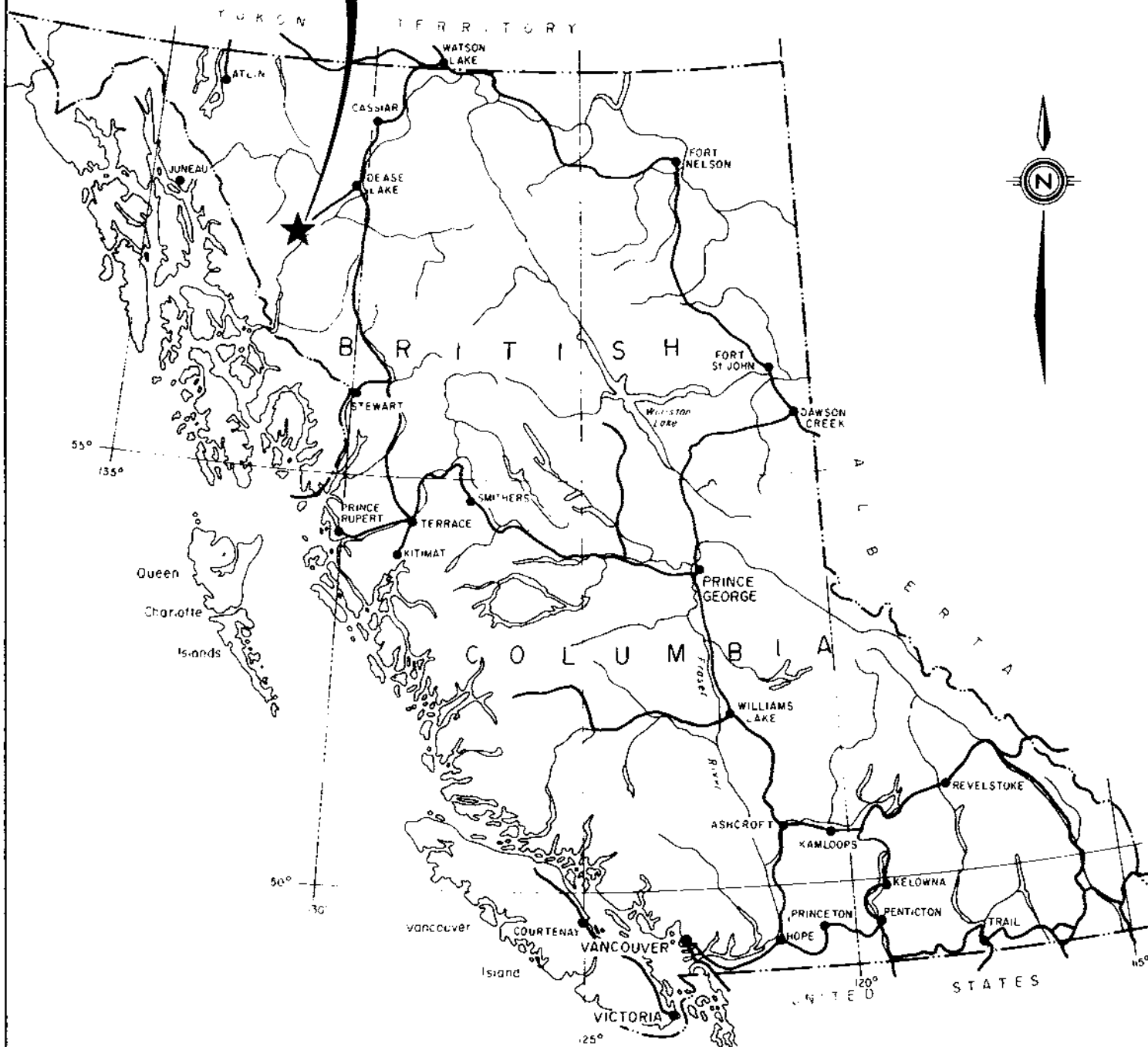
2.0 LIST OF CLAIMS

Records of the British Columbia Ministry of Energy, Mines and Petroleum Resources indicate that the following claims (Figure 2) are owned by Continental Gold Corp. Separate documents indicate that they are under option to Candela Resources Ltd.

Claim Name	Record Number	No. of Units	Record Date	Expiry Year
Shake 1	4695	20	June 27, 1988	1989
Shake 2	4696	20	June 27, 1988	1989
Shake 3	4697	20	June 27, 1988	1989
Shake 4	4698	20	June 27, 1988	1989
		80		

The position of the legal corner posts for the Shake 1-4 claims has not been verified by the author. The Shake 1-4 claims partially overlap the previously-staked Canyon 25 claim, resulting in the loss of almost 20 units.

PROPERTY LOCATION



CANDELA RESOURCES LTD.			
SHAKE 1-4 CLAIM GROUP			
PROPERTY LOCATION MAP			
EQUITY ENGINEERING LTD			
Drawn	NTS	Date	FIG. No
J.W.	104 G/13E	July, 1989	1.

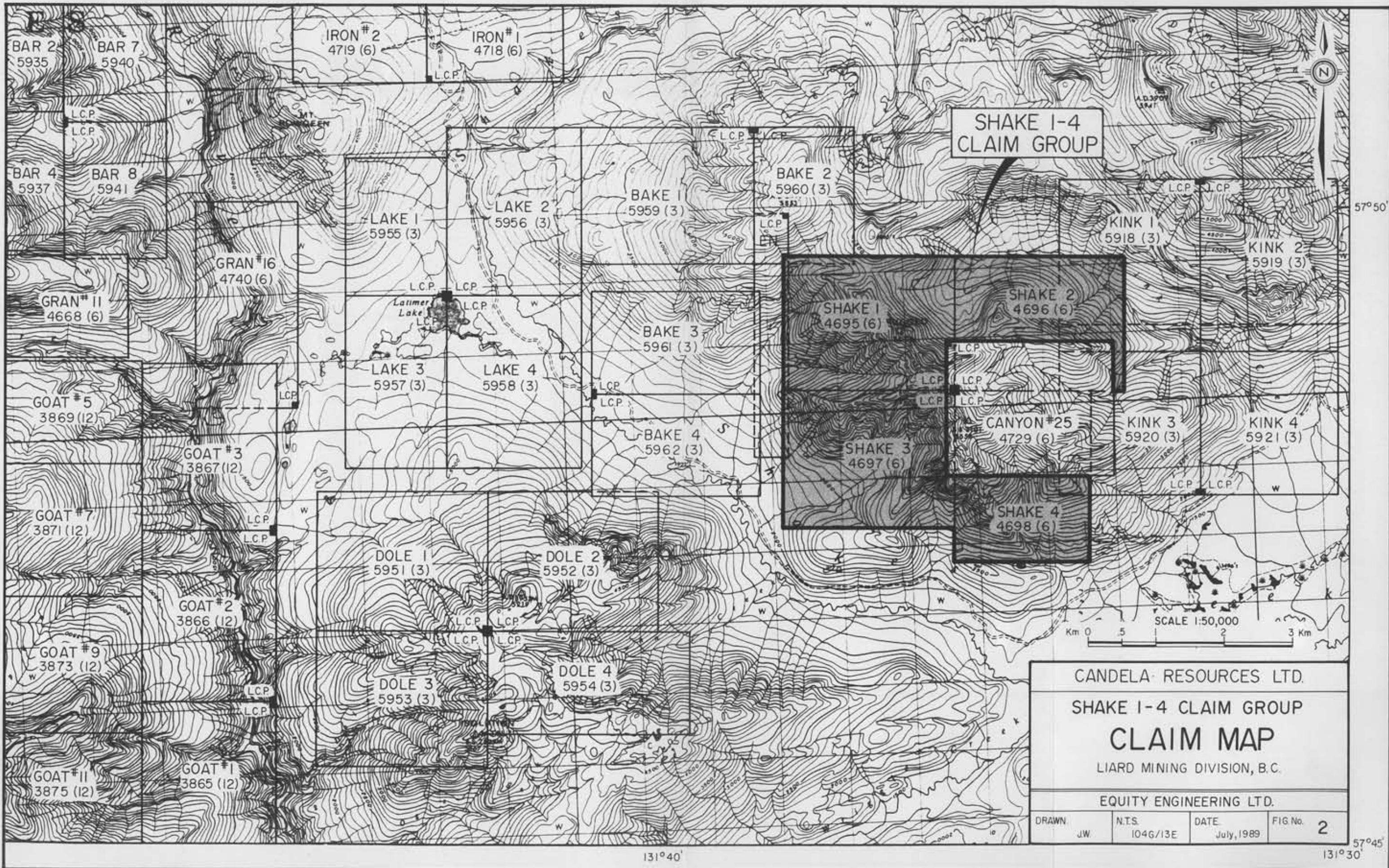
3.0 LOCATION, ACCESS AND GEOGRAPHY

The Shake 1-4 claims are located within the Coast Range Mountains approximately 28 kilometers southwest of Telegraph Creek in northwestern British Columbia (Figure 1). They lie within the Liard Mining Division, centered at 57° 49' north latitude and 131° 36° west longitude.

A secondary road extends sixteen kilometers south of Telegraph Creek to Glenora on the Stikine River. An access road suitable for four-wheel drive vehicles has been constructed southwest from Glenora to the site of a placer mining camp on the Barrington River, passing within ten kilometers of the Shake property. A cat road was built in the 1960's up Shake Creek from the Barrington River road, passing within a few hundred meters of the southwest corner of the Shake 3 claim. This cat road would have to be cleared and upgraded before it could be accessed. Access to the Shake property for the 1989 exploration program was provided by daily helicopter setouts from Glenora, a distance of less than fifteen kilometers.

The Shake 1-4 claims cover the higher elevations of Rugged Mountain (Figure 2). Topography is rugged, typical of mountainous and glaciated terrain, with elevations ranging from 635 meters on a tributary of Shake Creek termed Syenite Creek in this report to 1823 meters on the peak of Rugged Mountain.

Lower slopes are covered by a dense growth of coniferous growth including pine, hemlock and spruce with an undergrowth of devil's club and huckleberry. Steeper open slopes are covered by dense slide alder growth. Above treeline, which occurs between 1000 and 1400 meters elevation, more open alpine vegetation is present.



CANDELA RESOURCES LTD.
 SHAKE 1-4 CLAIM GROUP
CLAIM MAP
 LIARD MINING DIVISION, B.C.
 EQUITY ENGINEERING LTD.

DRAWN JW	N.T.S. 104G/13E	DATE July, 1989	FIG No. 2
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131°40'

57°45'
131°30'

The property lies in an intermediate or gradational belt between the wet belt of the Coast Range and the dry belt of the Stikine Plateau. There is little rain during the summer months and the snowfall is considerably lighter than in the wet belt. Prospecting could be started in May and continued through to the middle of October except on the highest elevations or on protected north-facing slopes.

4.0 PROPERTY MINING HISTORY

4.1 Previous Work

Placer gold was discovered on gravel bars of the Stikine River between Glenora and Telegraph Creek in 1861 and worked extensively until the early 1900's. The placer gold deposits of the lower Barrington River, ten kilometers southwest of the Shake 1-4 claims, have been worked sporadically since 1903.

The area south and west of Telegraph Creek was extensively explored for its copper potential throughout the 1960's, following the discovery of the Galore Creek copper-gold porphyry deposit in 1955 and the Schaft Creek copper-molybdenum deposit in 1957, both of which host greater than one million tonnes of contained copper. These deposits are located 85 kilometers south-southwest and 60 kilometers south, respectively, of Telegraph Creek.

Several copper occurrences were discovered southwest of Telegraph Creek at this time. Kennco explored copper mineralization within a syenite body and its intruded volcanics on their Poke claims, 13 kilometers west of Rugged Mountain (BCDM, 1963-65). Their Gordon claims, 10 kilometers west of Rugged Mountain at the junction of Limpoke Creek and the Barrington River,



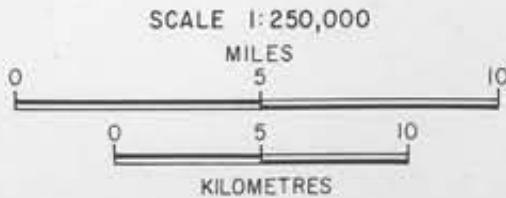
LEGEND

- QUATERNARY
PLEISTOCENE AND RECENT**
- 20 Fluvial gravels, sand, silt; glacial outwash, till, alpine moraine and colluvium
- TERTIARY AND QUATERNARY
UPPER TERTIARY AND PLEISTOCENE**
- 29 Basalt, olivine basalt, dacite-related pyroclastic rocks and subvolcanic intrusions; minor rhyolite; in part younger than some 28
- CRETACEOUS AND TERTIARY
UPPER CRETACEOUS AND LOWER TERTIARY
SLOGO GROUP**
- 24 Light green, purple and white rhyolite, trachyte and dacite flows, pyroclastic rocks and derived sediments
- SUNTUY GROUP**
- 21 Chert-pebble conglomerate, granite-boulder conglomerate, quartzose sandstone, arkose, siltstone, carbonaceous shale and minor coal
- 18 Medium-to coarse-grained, pink biotite-hornblende quartz monzonite
- JURASSIC AND/OR CRETACEOUS
POST-UPPER TRIASSIC PRE-TERTIARY**
- 16 Hornblende diorite
- 17 Granodiorite, quartz diorite; minor diorite, leucogranite and migmatite
- LOWER JURASSIC**
- 13 Conglomerate, polymictic conglomerate; granite-boulder conglomerate, grit, greywacke, siltstone; basaltic and andesitic volcanic rocks, peperite, pillow-breccia and derived volcanoclastic rocks
- TRIASSIC AND JURASSIC
POST-UPPER TRIASSIC PRE-LOWER JURASSIC**
- 12 Syenite, orthoclase porphyry, monzonite, pyroxenite
- TRIASSIC
UPPER TRIASSIC**
- 9 Undifferentiated volcanic and sedimentary rocks (units 5 to 9 inclusive)
- 8 Andesite-dacite flows, pyroclastic rocks, derived volcanoclastic rocks and related subvolcanic intrusions; minor greywacke, siltstone and polymictic conglomerate
- 7 Siltstone, thin-bedded siliceous siltstone, ribbon chert, calcareous and dolomitic siltstone, greywacke, volcanic conglomerate, and minor limestone
- 6 Limestone, fatid argillaceous limestone, calcareous shale and redbed limestone; may be in part younger than some 7 and 8
- PERMIAN
MIDDLE AND UPPER PERMIAN**
- 3 Limestone, thick-bedded mainly bioclastic limestone; minor siltstone, chert and tuff
- PERMIAN AND OLDER**
- 2 Phyllite, argillaceous quartzite, quartz-sericite schist, chlorite schist, gneiss, minor chert, schistose tuff and limestone
- 1 Amphibolite, amphibolite gneiss; age unknown probably pre-Upper Jurassic



SYMBOLS

- Geological boundary (defined and approximate, assumed) - - - - -
- Bedding (horizontal, inclined, vertical, overturned) + / / /
- Anticline + / / /
- Syncline - / / /
- Fault (defined and approximate, assumed) - - - - -
- Thrust fault, teeth on hanging-wall side (defined and approximate, assumed) - - - - -
- Fossil locality (circle with dot)
- Mineral property (circle with 'M')
- Glacier (wavy line)



CANDELA RESOURCES LTD.
 SHAKE 1-4 CLAIM GROUP
 REGIONAL GEOLOGY
 LIARD MINING DIVISION, B.C.

EQUITY ENGINEERING LTD.

DRAWN. J.W.	N.T.S. 1046/13E	DATE July, 1989	FIG. No. 3
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also hosted disseminated copper mineralization within syenite and the intruded volcanics (BCDM, 1966). The MH iron deposit, hosted by a pyroxenite stock on Shakes Creek seven kilometers northwest of Rugged Mountain, was also explored extensively in the 1960's.

No work has ever been recorded on the ground currently covered by the Shake 1-4 claims. However, claim tags dated June 18, 1971 were found immediately west of Rugged Mountain peak. The claims, staked on behalf of Amax Explorations, are part of the Horn group of claims that once covered Rugged Mountain in a northwesterly direction. The ground was undoubtedly acquired by Amax for the porphyry copper potential of the syenite body on the southern flank of Rugged Mountain.

4.2 1989 Work Program

During June of 1989, Candela Resources Ltd. carried out reconnaissance exploration on the Shake 1-4 claims, consisting of geological mapping, prospecting and stream sediment sampling, using a topographic orthophoto at a scale of 1:5000. This program was targeted at gold-rich mesothermal base metal veins similar to those occurring within a similar geological environment to the southeast in the Galore Creek, Iskut River, Sulphurets and Stewart mining districts.

During the course of this program, 10 silt samples and 57 rock samples were taken. Silt samples were taken from silt accumulations in major drainages, sieved to minus 80 mesh in the laboratory and analysed geochemically for gold and 32-element ICP (Figure 4). Prospecting and reconnaissance geological mapping were carried out over all claims, using a 1:5000 orthophoto with ten-meter contour intervals as a base (Figure 4). Rock samples, described in Appendix C, were taken from zones of alteration and mineralization and analysed geochemically for gold and 32-element

ICP. One sample containing greater than 2000 ppb gold was fire assayed for gold. Analytical certificates are attached in Appendix D.

5.0 REGIONAL GEOLOGY

The Telegraph Creek area lies on the western margin of the Intermontane Belt within the Stikine Arch near its contact with the Coast Plutonic Complex (Figure 3). A sequence of Paleozoic to middle Triassic oceanic sediments is unconformably overlain by Upper Triassic Stuhini Group island arc volcanics and sediments. These have been intruded by Upper Triassic to Lower Jurassic syenitic stocks and by Jurassic to Lower Cretaceous quartz diorite and granodiorite plutons of the Coast Plutonic Complex.

The oldest rock assemblage in the Telegraph Creek area consists of Permian bioclastic limestone (Unit 3) overlying metamorphosed sediments and volcanics (Unit 2) and crinoidal limestone (Unit 1).

Unconformably overlying the Permian limestone unit are Upper Triassic Stuhini Group island arc volcanics and sediments (Units 5 through 8). In the Telegraph Creek area, Souther (1971) grouped these volcanic and sedimentary members in Unit 9, noting however that it was composed predominantly of augite andesite breccia, conglomerate and volcanic sandstone. Several significant gold occurrences are hosted by Upper Triassic Stuhini volcanics in a cluster around Galore Creek seventy kilometers to the south. This Upper Triassic volcanosedimentary package is also correlative with that which hosts the Snip and Stonehouse gold deposits of the Iskut River district a further sixty kilometers to the south.

Small, equidimensional syenite, pyroxenite and orthoclase porphyry stocks (Unit 12), dated as Late Triassic to Early Jurassic by Souther (1971), intrude mainly Stuhini volcanics. Two of these stocks outcrop on Rugged Mountain and Mount Rowgeen. The syenite porphyry associated with the Poke and Gordon copper occurrences, ten kilometers west of Rugged Mountain, may also belong to Unit 12. The Galore Creek and Copper Canyon copper-gold porphyry deposits are also hosted by Upper Triassic volcanics intruded by syenitic stocks of Unit 12. Orthoclase porphyry or syenite stocks are associated with most significant precious metals deposits in the Stewart, Sulphurets and Iskut River districts, including the Silbak Premier, Sulphurets, and Snip deposits.

Lower Jurassic conglomerates (Unit 13) with granodiorite clasts unconformably overly Triassic sediments of the Stuhini Group. The Jurassic volcano-sedimentary strata are similar in appearance to those of the underlying Stuhini Group, with differentiation possible mainly through fossil identification.

Jurassic and Cretaceous granodiorite to quartz diorite batholiths (Unit 17) of the Coast Plutonic Complex intrude all older lithologies. This unit consists mainly of medium-grained hornblende-biotite granodiorite with lesser hornblende quartz diorite and is locally foliated near its margins.

Coarse conglomerate, sandstone, siltstone and minor black shale of the Upper Cretaceous and Lower Tertiary Sustut Group (Unit 21) unconformably overlies Jurassic strata on Mount Helveker and are found along the Stikine River below Telegraph Creek. Conformably overlying the Sustut Group on Helveker Mountain are about 160 meters of felsic to intermediate, mainly pyroclastic rocks (Unit 24), correlated by Souther (1972) to the Early Tertiary Sloko Group found further to the northwest.

Upper Tertiary and Quaternary basalt flows (Unit 25) are exposed in the Stikine River and north of Dodjatin Mountain.

6.0 PROPERTY GEOLOGY AND MINERALIZATION

6.1 Geology

Reconnaissance geological mapping indicates that the property is underlain by volcanic and sedimentary rocks of the Upper Triassic Stuhini Group (Unit 8) that are intruded by a differentiated syenitic intrusive (Unit 12) on the southern flank of Rugged Mountain (Figure 4). The legend on the property geology map was selected in order to incorporate the divisions and numbering used on Souther's 1971 Telegraph Creek map.

Volcanic members of the Stuhini Group consist of dark green, mafic volcanoclastics (Unit 8a) and minor augite porphyry flows. The volcanoclastic rocks are easily identified by the appearance of subrounded to subangular fragments on weathered surfaces. The clast size varies from a tuffaceous grit to small blocks, greater than ten centimeters across, characteristic of a volcanic breccia. The dark green color of the tuff/breccia is due to a pervasive chloritization. Exposures of this rock unit are found north of Contact Creek on the west side of the Shake 1 claim. On the south end of the claim group in Forelorn Creek, a bedded volcanic sandstone outcrops (Unit 8b). This rock is distinguished by layering one to four centimeters wide, and is very well jointed.

Overlying the volcanic group is a mixed sedimentary package of laminated siltstone/greywacke, argillite and thin discontinuous limestone horizons (Unit 7). The package characteristically weathers to a rusty colour, reflecting 1-5% pyrite content and this is exemplified by the gossans on the west side of Rugged Mountain

peak which are underlain by pyritic, thinly laminated, calcareous, siltstone/greywacke. Bedding measurements indicate these sediments strike northwesterly and dip in a shallow to moderate northeasterly direction. Fossil casts of pelecypod (Monotis?) can be found in argillite subcrop northwest of Rugged Mountain. This fossil would confer with the assigned Upper Triassic age of this sedimentary package. Souther (1971) has tentatively mapped Unit 7 as being conformably overlain by the volcanic rocks (unit 8); however, no bedding or structural evidence was seen to explain why the sedimentary succession is found overlying the volcanoclastic package in this area. The stratigraphic positioning exhibited on the Shake property agrees the stratigraphic section proposed by earlier government mapping by Kerr (1948).

Further west, close to the northern boundary of the Shake 1 claim is a polymictic conglomerate (unit 8c) containing siltstone, argillite, limestone, flower porphyry and volcanoclastic clasts. As the unit is fairly resistant to weathering, the conglomerate forms steep bluffs and surface weathering out of limestone clasts gives a pocked weathered surface. The clasts within this unit are well rounded and may be up ten centimeters in diameter.

The intrusive rocks (Unit 12) are separated into three categories: syenite, orthoclase porphyry and pyroxenite. The first category forms the main plutonic mass in the southwest quadrant of the property whereas the second two rock types are related to the main syenite by a high content of potash feldspar and an absence of quartz. This group of alkaline intrusives cut Upper Triassic volcanic and sedimentary rocks and are dated as late Triassic to early Jurassic.

The main body on Rugged Mountain is a mainly equigranular, locally porphyritic syenite (Unit 12a) consisting of orthoclase and minor mafic minerals. The colour of this unit varies from a grey

colour to reddish pink. Kerr (1948) differentiated the two types on the basis of colour and cross-cutting relationships leading him to conclude that "the red colour is due to the development, or introduction, of the unknown reddening constituents." The grey syenite is predominately on the north side of the pluton and the red syenite is more to the south and can be found cutting the grey mass. The "unknown reddening agent" proposed by Kerr is probably an alteration product of potassium feldspar introduced late in the crystallization of Rugged Mountain syenite. Dykes of the syenite material are found in the surrounding volcanic rocks near to the contact of the main mass.

A rind of dark green, pyroxenite (Unit 12c) occurs on the syenite-volcanic/sedimentary contact. The pyroxenite is comprised of a high content of granular, disseminated magnetite (up to 50%), biotite and a dark green pyroxene, augite. The pyroxenite exhibits sharp contact boundaries with both the syenite and the surrounding volcanic rocks following the length of Contact Creek. The formation of the pyroxenite may be a result of the assimilation of mafic volcanic material and perhaps, by the differentiation of the main syenitic body. The pyroxenite shell on Rugged Mountain appears to be very similar to the magnetite-rich pyroxenite described on the MH iron property located northwest of the Shake property.

The third intrusive type (Unit 12b) is that of the orthoclase porphyry dykes (sills?) found crossing Rugged Mountain in a northeasterly direction. These dykes are characterized by large orthoclase phenocrysts up to several centimeters across, set within a light grey, aphanitic groundmass. The dykes are generally not greater than 10 meters wide and contacts with the country rock are sharp.

6.2 Mineralization

Two types of sulphide mineralization were discovered during the current exploration program: shear hosted, quartz-carbonate veining and syenite hosted veining and fracture fillings. The greatest precious metal potential exists in the veining of the first type.

The strongest sulphide mineralization and best gold values were found in narrow quartz-carbonate veins and discontinuous massive sulphide lenses within shear structures in altered volcanic and sedimentary rocks north of Contact Creek. The mineralogy consists mainly of pyrite, chalcopyrite, magnetite and arsenopyrite with as well as a single occurrence of molybdenite. These zones of mineralization are easily identified by their rusty weathered surfaces of limonitic products and copper staining. Syenite and orthoclase dykes are quite often found in close proximity to these mineral occurrences. Although some of the structures are well mineralized, the majority of them are limited in both width and strike length. Typical of these structures is a narrow sheared zone with isolated pods of pyrite and chalcopyrite, sampled north of Contact Creek (Sample# 172302). A select grab of the sulphide pods contained 790 parts per billion gold and 55.0 parts per million silver.

The highest gold values were returned from a 5 to 20 centimeter wide quartz-carbonate vein, mineralized with pyrite, arsenopyrite and chalcopyrite, that can be traced for approximately 50 meters. The vein is hosted in pyritic sediments and strikes northeasterly and dips steeply to the southeast. Sample# 30268, a grab sample from the zone, assayed 0.422 ounces per ton gold.

The syenite was found to contain chalcopyrite and pyrite mineralization along fractures and in veins, more typical of

porphyry style mineralization. One of the vein structures was found to be anomalous in gold with a geochemical value of 1120 parts per billion (Sample# 172308). The frequency and density of the mineralized structures do not appear great enough in the areas examined to host a large tonnage, porphyry style deposit.

Character samples were taken of the magnetite rich pyroxenite surrounding the main syenite body (Sample# 172304) and weakly pyritic orthoclase porphyry dykes (Sample# 172305). Gold values for both rock types were below detection limit and the copper content of the pyroxenite was weakly anomalous at 533 parts per million.

7.0 STREAM GEOCHEMISTRY

Ten silt samples were taken from streams draining the Shake 1-4 claims (figure 5). The results confirm, in part, the results of the National Reconnaissance survey released in 1988 and reflect the mineralization found north of Contact Creek in the altered volcanic and sedimentary rocks.

The government survey showed the drainages on the west side of Rugged Mountain (Syenite, Contact, Forelorn, North Creeks) to be anomalous in copper, cobalt and gold.

Eight of the silt samples taken this year were very anomalous in copper (up to 720 parts per million) reflecting the copper mineralization in the syenite, volcanic and sedimentary rocks on the west side of the Shake property. Four of the samples contained elevated gold values ranging from 40 to 100 parts per billion. Three of these samples were taken from northern tributaries of Contact Creek in areas underlain by altered volcanics and sediments intruded by syenite and orthoclase porphyry dykes. This area has

been identified by mapping and prospecting to be a favourable host for precious metal mineralization and this is being reflected in the anomalous stream geochemistry.

One sample taken (S-8) on a tributary of Syenite Creek returned a gold value of 45 parts per billion. The source of this anomaly was found up slope in a narrow quartz-carbonate vein with pyrite and chalcopyrite. A grab sample of this vein, Sample# 172308, contained 1120 parts per billion gold.

8.0 DISCUSSION AND CONCLUSIONS

During the course of limited exploration work in 1989, anomalous stream sediment and rock samples were taken from the western slope of Rugged Mountain on the Shake 1-4 mineral claims.


Rock samples collected from the property which showed notable gold and copper geochemistry were taken from a number of shear vein structures hosted in volcanic and sedimentary rocks in close proximity to a syenite intrusion and satellite orthoclase porphyry and syenite dykes. The narrow, quartz-carbonate veins contain pyrite, chalcopyrite, arsenopyrite and magnetite. To date, the best results come from a 5 to 20 centimeter wide vein with arsenopyrite mineralization. A grab sample from this vein assayed 0.422 ounces per ton gold.

Silt sampling on the Shake 1 claim indicated that three of the drainages contained gold values of 40, 90 and 100 parts per billion with very anomalous copper values. The source of most of these anomalies remain to be discovered.

Extremely encouraging initial results, coupled with the exploration successes achieved all along the regional trend between

the Stewart, Iskut River, Galore Creek and Schaft Creek areas provide abundant incentive to conduct further exploration work on the Shake 1-4 mineral claims.

Respectfully submitted,
EQUITY ENGINEERING LTD.


David A. Caulfield, F.G.A.C.

Vancouver, British Columbia
September, 1989

APPENDIX A

BIBLIOGRAPHY

BIBLIOGRAPHY

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APPENDIX B

STATEMENT OF EXPENDITURES

STATEMENT OF EXPENDITURES
SHAKE 1-4 CLAIMS
(June 16-23, 1989)

PROFESSIONAL FEES AND WAGES:

David A. Caulfield, Geologist		
7.75 days @ \$350/day	\$	2,712.50
Henry J. Awmack, P.Eng.		
0.5 days @ \$350/day		175.00
Kika Ross, Geologist		
2.5 days @ \$250/day		625.00
Tom Bell, Prospector		
6.5 days @ \$250/day		1,625.00
Don Coolidge, Prospector		
6.0 days @ \$250/day		1,500.00
Clerical		
12 hours @ \$20/hour		<u>240.00</u>
		\$ 6,877.50

EQUIPMENT RENTALS:

Truck		
3.5 days @ \$60/day	\$	210.00
Truck (Standby)		
3.0 days @ \$10/day		30.00
Hand-held Radios		
10 days @ \$5/day		50.00
Fly Camp		
21 mandays @ \$20/manday		<u>420.00</u>
		710.00

CHEMICAL ANALYSES:

10 silt @ \$13.87	\$	138.65
57 rock geochem @ \$16.45		937.65
1 gold assay @ \$8.46		<u>8.46</u>
		1,084.76

EXPENSES:

Materials and Supplies	\$	164.90
Maps and Publications		32.31
Orthophoto Construction	3,206.00	
Printing and Reproductions		105.69
Camp Food		354.50
Camp Fuel		11.95
Camp Supplies		48.12
Meals		80.08
Travel		458.30
Automotive Fuel		99.95
Aircraft Charters		339.00
Helicopter Charters	3,359.40	
Telephone Distance Charges		151.85
Freight		81.00
Expediting		18.00
Courier and Telefax		<u>40.36</u>
		8,551.41

REPORT (estimated)

2,000.00

MANAGEMENT FEE:
15% on expenses

1,741.45

\$ 20,965.12
=====

APPENDIX C

ROCK DESCRIPTIONS

Sampler Tom Bell

Project CDD89-02

Location Ref Rugged Mtn

Date June 19-22, 1989

Property Shakes 1-4

Air Photo No _____

SAMPLE NO.	LOCATION	SAMPLE TYPE	Sample Width	True Width	DESCRIPTION			ADDITIONAL OBSERVATIONS	ASSAYS					
					Rock Type	Alteration	Mineralization		ppb Au	ppm Ag	ppm Cu	ppm Pb	ppm Zn	ppm As
30267	6411650 N 343350 E Creek North of Contact Creek	Grab DC	5.0m	2.0m	mafic Volc.	SE, LI	PY, MN, Mg tr. MO	shear zone $\pm 120^\circ/90^\circ$ NE sample of HW side	15	1.0	824	<2	42	40
58	6411650 N 343370 E	Grab DC	3.0m	"	"	"	"	same zone 10m up slope. zone exposed 15m	35	1.2	1220	12	42	55
59	6411510 N 343570 E (1220m Elev.)	Grab DC	50cm	1-2m	Contact w/ syenite dike	CA, AK	LI, PY	$\pm 065^\circ?$, gossan extends across face for 50m	20	0.6	523	8	62	60
60	6411500 N 343550 N	Subcrop Grab	10cm	?	"	AK	PY	bottom end of above gossan on another dike	40	<0.2	280	2	16	25
61	6411610 N 343610 E	Grab DC	50cm	50cm	mafic volcanics	QE, CA	PY	west side of gully, 50x50m pod	60	0.6	924	<2	114	25
62	6411670 N 343690 E (1275m Elev.)	Float			?	CA, GA, EP	PY (coarse euhedral)		15	0.2	134	2	26	10
63	6411550 N 343350 E (1120m Elev.)	Grab DC	25cm	50cm	alt & volcanics	CA	LI, PY, MN	$\pm 100^\circ/V$, zone exposed for 5.0m	20	1.4	342	<2	62	30
64	6411510 N 343300 E (1200m Elev.)	Grab DC	1.0m	1.0m	Contact of lime, volc. w/ alt. porphy	CL, CA	LI, PY, CP AZ, MA, HN	$\pm 080^\circ/V$	25	1.2	409	10	66	15
65	6411350 N 344720 E Between Contact & North Creek	Grab DC	1.0m	10-15cm	volcanics	SE, CL, QE	PY, AR, CP, ZN	1720m Elev., Gossan, exposed 30-40m $\pm 025^\circ/75^\circ$ SE, grab along vein	730	0.4	80	14	48	710,000
66	6411350 N 344720 E	Grab DC	2.0m	5-10cm	"	CA, SE	PY, AR	10m along vein, grab for 2m along strike - east of 65	5370	2.0	178	24	66	8500
67	6411350 N 344720 E	Grab DC	50cm	5cm	siltst.	CA, SE	PY, AR	6m west of 65	370	<0.2	54	6	32	5120
68	6411350 N 344720 E	Grab DC	5m	5-10cm	"	CA, SE	PY, AR	15m-20m west of 65, vein split	710,000	6.8	80	14	40	5300
69	6411350 N 344720 E	Grab DC	20cm	50x50cm	laminated siltst.	CL	PY, GE, JA CP	pod of mineralization off of CA vein between 267, 268	200	1.4	993	2	50	105
70	6411300 N 344660 E (1665m)	Grab DC	1.0m	5-10cm	"	CA, QE	AR, PY	same system as 265 down slope $\pm 040^\circ/V$	375	0.2	23	32	30	1495
71	644300 N 344660 E	Grab DC		5-10cm	"	CA, QE	AR, PY, GA	3m east of 270, 2nd vein $030^\circ/V$	180	<0.2	15	48	106	1315
72	6411170 N 344750 E (1660m)	Grab DC	1.0m	50-60cm	orthoclase porphyry	SE, CL, CA	PY, GA, AZ MA	zone or pod?	20	0.8	23	468	720	40
73	6411170 N 344660 E	Grab DC	50cm		"	QE, CL, CA	PY, GE, JA	opposite side of porphyry 10m x 30m gossan	40	0.8	133	12	54	30
74	6411010 N 344740 E (1630m)	Grab DC	3.0m		alt & siltst.	CL, white chalky-CA	GE, TA, HE NE	south along face, area 5m x 5m	35	0.6	725	<2	82	60
75	6411140 N 344610 E (1575m)	Grab DC	1.0m	1.0x30m	alt & syenite	CL	JA, GE, PY gouge		680	1.8	338	6	40	35
76	6411140 N 344540 E (1520m)	Grab DC	2.0m	20-100cm	volcanics agglon.	CA, QE, CL	PY, JA, GE HE	12m strike, $\pm 100^\circ/V$	50	0.2	318	14	186	40

Sampler Don Coolidge

Project CDD 89-02

Location Ref Rugged Men.

Date June 19-22, 1989

Property Shakes 1-4

Air Photo No _____

SAMPLE NO.	LOCATION	SAMPLE TYPE	Sample Width	True Width	DESCRIPTION			ADDITIONAL OBSERVATIONS	ASSAYS					
					Rock Type	Alteration	Mineralization		ppb	ppm	Au	Ag	Cu	Pb
358004	6410830 N 343560 E Contact Creek	Float grab			mafic volc.	CL	PY, MG, MA	float in talus slope	30	0.2	319	28	66	5
05	6411010 N 343680 E (1273m)	Float grab			Syenite porphyry	CL, EF	PY, MO, FMO	talus, mineralized in volc. in contact w/ dike	25	1.2	484	22	112	<5
06	6411000 N 343770 E (1183m)	Grab dc	10cm		mafic volc.	CL	LI, PY	1cm lense of massive PY small pod.	<5	0.6	1725	8	90	5
07	6411000 N 343770 E (1180m)	Grab dc	4cm		"	QE		+020°/80°SE, sugary texture QE stringer	260	<0.2	44	6	12	35
08	6410980 N 343690 E (1220m)	Grab dc	90cm		syenite porphyry	CL	CP, MG, PY	weather orange to tan 103°/?, slickensides on comb.	10	0.4	863	10	36	<5
09	6410900 N 344230 E (1290m)	Grab dc	12cm		mafic volc.	CL	PY	Rusty zone 12cm wide, small pod, SY.P in area	25	0.8	1585	12	84	10
10	6410820 N 344210 E (1290m)	Grab dc	10x14cm		"	CL	PY, MA, CP	Possibly on strike w/ 09 5m to south.	125	4.2	7910	8	94	35
11	6410860 N 344200 E (1100m)	Grab dc	14-20cm		"	CL		+120° trend, 1m. strike length only	50	1.2	3390	6	58	15
12	6411360 N 344720 E (1775m) Rugged Men. Peak	Grab dc			interbedded silt/sandstone		PY, LI	Rusty stained dc. adjacent to orthoclase porphyry	20	<0.2	201	6	30	5
13	6411570 N 344850 E (1619m)	Grab dc	1-3m		Volc.	CL	PY, LI	greenish volc. w/ disse PY exposed 25m trend 10-20°	<5	<0.2	189	4	20	<5
14	6407760 N 346890 E (710m) (along creek)	Chip	17m	~17m	well alt'd shattered	CL, CA	PY, JA, GE	} two consecutive samples of shattered, alt'd volcaniclastic in contact w/ syenite	<5	<0.2	292	<2	52	15
15	6407780 N 346870 E (710m)	Chip	20m	~20m	volcaniclastic	"	"		<5	<0.2	302	<2	44	15
16	6407810 N 346700 E (774m)	Grab dc	18x20cm		"	CL	PY, GE	sample 50m above creek concentrated pod of 5"	<5	<0.2	226	<2	30	<5
17	6407810 N 346900 E (774m)	Grab dc	50x60cm		"	CL	PY, MG, CP	10m west of 358016, taken adjacent to dyke ± 010°	<5	<0.2	384	12	32	5
18	6408090 N 346610 E (793m)	Grab dc	12x16cm		syenite		PY 2-5%	sample 35m above creek sample pod	<5	<0.2	584	6	52	135
19	6408130 N 346550 E (823m)	Grab dc	0.5m		alt'd volcaniclastic	well alt'd CL, SE	PY	25m above creek in mud slide area, ± 084° for 1.5m	25	<0.2	128	8	24	45
20	6410970 N 344950 E Contact Cr.	Grab dc	40cm		alt'd sediments	EP?	PY, CP, HE? LI,	Rusty mudst. dc, ± 026° for 40m.	80	<0.2	699	4	42	30
21	6410970 N 344950 E (1710m)	Chip dc	20cm		"	CL	15-20% PY JA, GE	20m south of 020, gongle on FW, blob 20x35cm	85	<0.2	1140	<2	48	15
22	6410970 N 345010 E (1775m)	Grab dc	40cm		alt'd orth. porphy		GE	almost ferricite appearance all 5" weathered out	45	<0.2	311	12	280	40
23	6410970 N 345010 E (1775m)	Grab dc	4-8cm		"	CA	no S"	strikes 20m 038°/88°E	<5	<0.2	6	4	10	5

APPENDIX D

CERTIFICATES OF ANALYSIS



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
 212 BROOKSBANK AVE., NORTH VANCOUVER,
 BRITISH COLUMBIA, CANADA V7J-2C1
 PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
 V6B 1N2

A8919181

Comments: ATTN: D. CAULFIELD

CERTIFICATE A8919181

EQUITY ENGINEERING LTD.

PROJECT : CDD89-02

P.O.# : NONE

Samples submitted to our lab in Vancouver, BC.

This report was printed on 9-JUL-89.

SAMPLE PREPARATION

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION
205	57	Rock Geochem: Crush, splitting
238	57	ICP: Aqua regia digestion

• NOTE 1:

The 32 element ICP package is suitable for trace metals in soil and rock samples. Elements for which the nitric-aqua regia digestion is possibly incomplete are: Al, Ba, Be, Ca, Cr, Ga, K, La, Mg, Na, Sr, Ti, Tl, W.

ANALYTICAL PROCEDURES

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	UPPER LIMIT
100	57	Au ppb: Fuse 10 g sample	FA-AAS	5	10000
921	57	Al %: 32 element, soil & rock	ICP-AES	0.01	15.00
922	57	Ag ppm: 32 element, soil & rock	ICP-AES	0.2	200
923	57	As ppm: 32 element, soil & rock	ICP-AES	5	10000
924	57	Ba ppm: 32 element, soil & rock	ICP-AES	10	10000
925	57	Be ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
926	57	Bi ppm: 32 element, soil & rock	ICP-AES	2	10000
927	57	Ca %: 32 element, soil & rock	ICP-AES	0.01	15.00
928	57	Cd ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
929	57	Co ppm: 32 element, soil & rock	ICP-AES	1	10000
930	57	Cr ppm: 32 element, soil & rock	ICP-AES	1	10000
931	57	Cu ppm: 32 element, soil & rock	ICP-AES	1	10000
932	57	Fe %: 32 element, soil & rock	ICP-AES	0.01	15.00
933	57	Ga ppm: 32 element, soil & rock	ICP-AES	10	10000
934	57	Hg ppm: 32 element, soil & rock	ICP-AES	1	10000
935	57	K %: 32 element, soil & rock	ICP-AES	0.01	10.00
936	57	La ppm: 32 element, soil & rock	ICP-AES	10	10000
937	57	Mg %: 32 element, soil & rock	ICP-AES	0.01	15.00
938	57	Mn ppm: 32 element, soil & rock	ICP-AES	5	10000
939	57	Mo ppm: 32 element, soil & rock	ICP-AES	1	10000
940	57	Na %: 32 element, soil & rock	ICP-AES	0.01	5.00
941	57	Ni ppm: 32 element, soil & rock	ICP-AES	1	10000
942	57	P ppm: 32 element, soil & rock	ICP-AES	10	10000
943	57	Pb ppm: 32 element, soil & rock	ICP-AES	2	10000
944	57	Sb ppm: 32 element, soil & rock	ICP-AES	5	10000
945	57	Sc ppm: 32 elements, soil & rock	ICP-AES	1	100000
946	57	Sr ppm: 32 element, soil & rock	ICP-AES	1	10000
947	57	Ti %: 32 element, soil & rock	ICP-AES	0.01	5.00
948	57	Tl ppm: 32 element, soil & rock	ICP-AES	10	10000
949	57	U ppm: 32 element, soil & rock	ICP-AES	10	10000
950	57	V ppm: 32 element, soil & rock	ICP-AES	1	10000
951	57	W ppm: 32 element, soil & rock	ICP-AES	10	10000
952	57	Zn ppm: 32 element, soil & rock	ICP-AES	2	10000



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Page No. : 1-A

Tot. Pages: 2

Date : 9-JUL-89

Invoice #: I-8919181

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8919181

SAMPLE DESCRIPTION	PREP CODE	Au ppb FA+AA	Al %	Ag ppm	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %	Mn ppm
30257 E	205 238	15	1.92	1.0	40	10	< 0.5	< 2	1.46	< 0.5	55	11	824	9.81	20	< 1	0.09	10	1.10	485
30258 E	205 238	35	3.93	1.2	55	10	< 0.5	< 2	3.90	< 0.5	78	12	1220	12.25	20	< 1	0.02	< 10	0.82	360
30259 E	205 238	20	2.09	0.6	60	60	< 0.5	< 2	1.04	< 0.5	31	11	523	11.95	20	< 1	0.08	20	0.75	555
30260 E	205 238	40	0.10	< 0.2	25	10	< 0.5	4	>15.00	< 0.5	18	< 1	280	2.05	< 10	< 1	0.02	< 10	0.12	1065
30261 E	205 238	60	2.07	0.6	25	10	< 0.5	< 2	9.32	< 0.5	40	75	924	9.81	10	< 1	0.08	< 10	2.25	1080
30262 E	205 238	15	1.72	0.2	10	< 10	< 0.5	< 2	10.80	< 0.5	48	10	134	5.07	< 10	< 1	< 0.01	< 10	0.77	705
30263 E	205 238	20	2.00	1.4	30	10	< 0.5	< 2	0.89	< 0.5	17	26	342	9.36	20	< 1	0.11	10	1.41	365
30264 E	205 238	25	1.83	1.2	15	20	2.0	< 2	2.74	< 0.5	26	11	409	5.02	10	< 1	0.06	10	0.68	895
30265 E	205 238	730	0.89	0.4	>10000	< 10	< 0.5	< 2	13.55	< 0.5	15	44	80	2.88	< 10	< 1	< 0.01	< 10	1.20	695
30266 E	205 238	5370	1.52	2.0	8500	10	< 0.5	< 2	10.90	< 0.5	19	51	178	3.52	< 10	< 1	0.15	< 10	1.21	680
30267 E	205 238	370	1.42	< 0.2	5120	10	0.5	< 2	12.10	< 0.5	12	53	54	2.95	< 10	< 1	0.03	< 10	1.83	885
30268 E	205 238	>10000	1.07	6.8	5300	< 10	0.5	< 2	13.15	< 0.5	13	39	80	2.56	< 10	< 1	0.08	< 10	1.16	795
30269 E	205 238	200	1.99	1.4	105	10	1.5	< 2	1.74	< 0.5	107	16	993	10.70	10	< 1	0.10	20	0.73	240
30270 E	205 238	375	0.64	0.2	1495	< 10	0.5	2	>15.00	< 0.5	6	36	23	1.54	< 10	< 1	0.06	< 10	0.62	1440
30271 E	205 238	180	0.60	< 0.2	1315	< 10	0.5	2	>15.00	< 0.5	4	27	15	1.10	< 10	< 1	0.15	< 10	0.53	1250
30272 E	205 238	20	0.90	0.8	40	10	1.5	< 2	7.52	9.0	7	13	23	7.71	< 10	< 1	0.07	< 10	0.86	1895
30273 E	205 238	40	0.90	0.8	30	< 10	1.0	< 2	0.10	< 0.5	25	18	133	5.20	10	< 1	0.04	< 10	0.47	305
30274 E	205 238	35	3.62	0.6	60	10	2.5	< 2	2.52	< 0.5	85	44	725	3.86	10	< 1	0.07	20	1.13	1740
30275 E	205 238	680	1.16	1.8	35	10	1.5	< 2	0.85	< 0.5	18	11	338	6.43	10	< 1	0.04	10	0.50	335
30276 E	205 238	50	3.52	0.2	40	< 10	2.0	< 2	11.30	0.5	40	18	318	8.29	< 10	< 1	0.09	< 10	0.77	1420
30277 E	205 238	35	3.15	< 0.2	25	10	< 0.5	< 2	4.32	8.0	65	24	79	9.25	10	< 1	0.12	< 10	1.78	1190
30278 E	205 238	55	1.84	< 0.2	15	10	< 0.5	< 2	1.80	< 0.5	42	21	136	14.15	20	< 1	0.08	10	1.25	915
172301 H	205 238	25	1.66	< 0.2	5	130	< 0.5	< 2	2.01	< 0.5	30	38	855	6.17	10	< 1	0.19	10	1.15	590
172302 H	205 238	790	2.56	55.0	60	10	< 0.5	< 2	2.06	9.5	84	15	>10000	13.15	20	< 1	0.06	10	2.13	845
172303 H	205 238	45	2.43	1.2	30	30	< 0.5	< 2	0.28	< 0.5	34	51	1095	8.51	10	< 1	0.64	< 10	2.82	670
172304 H	205 238	< 5	0.91	< 0.2	10	200	< 0.5	< 2	4.42	< 0.5	37	29	533	10.70	10	< 1	0.26	< 10	1.41	720
172305 H	205 238	< 5	0.81	< 0.2	10	20	< 0.5	< 2	2.27	< 0.5	5	7	69	1.92	< 10	< 1	0.20	20	0.29	595
172306 H	205 238	5	4.54	< 0.2	30	20	< 0.5	< 2	5.66	< 0.5	28	8	165	8.64	10	4	0.27	< 10	1.64	740
172307 H	205 238	210	1.66	9.8	10	10	< 0.5	< 2	6.48	< 0.5	33	22	3980	8.73	< 10	< 1	0.11	< 10	1.55	1615
172308 H	205 238	1120	0.44	2.0	20	70	< 0.5	< 2	0.17	< 0.5	8	8	127	7.19	< 10	< 1	0.29	< 10	0.07	150
172309 H	205 238	95	0.66	3.2	35	40	< 0.5	< 2	0.28	< 0.5	15	144	1010	>15.00	< 10	< 1	0.23	< 10	0.45	125
172310 H	205 238	60	0.34	1.4	10	100	< 0.5	< 2	0.33	< 0.5	5	53	242	6.50	< 10	< 1	0.30	< 10	0.25	90
172311 H	205 238	120	0.34	2.8	20	150	< 0.5	< 2	0.36	< 0.5	11	54	477	10.40	< 10	< 1	0.27	10	0.14	110
358004 H	205 238	30	2.41	0.2	5	70	< 0.5	< 2	2.54	< 0.5	25	50	319	5.40	< 10	< 1	0.38	10	1.67	750
358005 H	205 238	< 5	1.73	1.2	< 5	70	< 0.5	2	1.32	0.5	25	47	484	4.45	< 10	< 1	0.22	10	1.40	665
358006 H	205 238	< 5	1.68	0.6	5	30	< 0.5	< 2	2.00	< 0.5	13	59	1725	7.33	< 10	< 1	0.25	20	1.45	890
358007 H	205 238	260	0.33	< 0.2	35	< 10	< 0.5	< 2	5.40	< 0.5	4	74	44	1.05	< 10	< 1	< 0.01	< 10	0.32	460
358008 H	205 238	10	1.06	0.4	< 5	10	< 0.5	< 2	3.19	< 0.5	12	30	863	5.16	< 10	< 1	0.08	< 10	0.61	465
358009 H	205 238	25	2.51	0.8	10	30	< 0.5	< 2	2.72	< 0.5	57	53	1585	7.23	< 10	< 1	0.14	10	1.66	695
358010 H	205 238	125	2.10	4.2	35	10	< 0.5	< 2	3.78	< 0.5	118	25	7910	12.95	< 10	< 1	0.04	< 10	0.93	735

CERTIFICATION :

B. Caulfield



Chemex Labs Ltd.

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212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1

PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

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Page No. : 1-B
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CERTIFICATE OF ANALYSIS A8919181

SAMPLE DESCRIPTION	PREP CODE	Mo ppm	Na %	Ni ppm	P ppm	Pb ppm	Sb ppm	Sc ppm	Sr ppm	Ti %	Tl ppm	U ppm	V ppm	W ppm	Zn ppm
30257 E	205 238	1	0.02	12	1170	< 2	< 5	9	18	0.23	< 10	< 10	145	< 10	42
30258 E	205 238	10	0.01	15	1080	12	5	12	19	0.25	< 10	< 10	211	< 10	42
30259 E	205 238	2	0.02	4	4560	8	5	8	55	0.21	< 10	< 10	191	< 10	62
30260 E	205 238	109	0.01	2	310	2	< 5	1	1195	< 0.01	< 10	< 10	9	< 10	16
30261 E	205 238	< 1	0.01	26	1330	< 2	5	8	299	0.16	< 10	< 10	147	< 10	114
30262 E	205 238	11	0.01	57	310	2	< 5	2	584	0.03	< 10	< 10	81	< 10	26
30263 E	205 238	93	0.01	9	1120	< 2	< 5	10	137	0.37	< 10	< 10	196	< 10	62
30264 E	205 238	17	0.02	1	1090	10	5	5	71	0.12	< 10	< 10	137	< 10	66
30265 E	205 238	2	0.03	15	820	14	15	9	171	0.07	< 10	< 10	107	< 10	48
30266 E	205 238	7	0.03	21	890	24	10	12	138	0.12	< 10	< 10	101	< 10	66
30267 E	205 238	< 1	0.04	21	1080	6	10	12	155	0.18	< 10	< 10	131	< 10	32
30268 E	205 238	1	0.02	14	790	14	10	9	202	0.11	< 10	< 10	102	< 10	40
30269 E	205 238	2	0.07	33	1570	2	5	5	82	0.19	< 10	< 10	105	< 10	50
30270 E	205 238	< 1	0.01	5	340	32	5	3	771	0.02	< 10	< 10	37	< 10	30
30271 E	205 238	< 1	< 0.01	5	140	48	5	2	515	0.01	< 10	< 10	17	< 10	106
30272 E	205 238	< 1	0.01	3	60	468	5	1	101	0.01	< 10	< 10	57	< 10	720
30273 E	205 238	2	0.07	2	130	12	< 5	< 1	10	0.01	< 10	< 10	105	< 10	54
30274 E	205 238	7	0.03	81	1470	< 2	< 5	13	43	0.13	< 10	< 10	127	< 10	82
30275 E	205 238	10	0.03	5	1080	6	5	4	134	0.03	< 10	< 10	142	< 10	40
30276 E	205 238	19	0.01	7	570	14	5	7	93	0.13	< 10	< 10	106	< 10	186
30277 E	205 238	251	0.02	14	730	6	5	14	50	0.27	< 10	< 10	155	< 10	986
30278 E	205 238	272	0.01	4	640	14	5	14	17	0.21	< 10	< 10	161	< 10	106
172301 H	205 238	2	0.02	17	2300	< 2	< 5	4	244	0.25	< 10	< 10	252	< 10	72
172302 H	205 238	25	< 0.01	15	710	10	5	12	30	0.20	< 10	< 10	207	< 10	522
172303 H	205 238	7	< 0.01	17	1260	< 2	< 5	15	27	0.10	< 10	< 10	133	< 10	114
172304 H	205 238	< 1	0.03	10	6420	12	10	8	299	0.20	< 10	< 10	357	< 10	64
172305 H	205 238	1	0.04	< 1	630	18	< 5	1	154	0.02	< 10	< 10	65	< 10	80
172306 H	205 238	< 1	0.11	10	4400	< 2	5	13	90	0.32	< 10	< 10	335	< 10	84
172307 H	205 238	< 1	0.02	11	3780	24	5	24	341	0.78	< 10	< 10	549	10	124
172308 H	205 238	59	0.01	2	710	28	< 5	2	56	0.02	< 10	< 10	97	< 10	32
172309 H	205 238	4	0.01	21	960	12	5	19	28	0.93	< 10	< 10	258	< 10	42
172310 H	205 238	22	0.02	8	680	8	< 5	9	51	0.86	< 10	< 10	150	< 10	20
172311 H	205 238	17	0.01	28	1710	12	5	7	59	0.76	< 10	< 10	172	< 10	28
358004 H	205 238	1	0.26	21	1290	28	< 5	9	30	0.34	< 10	< 10	209	< 10	66
358005 H	205 238	>10000	0.01	16	960	< 2	5	11	91	0.04	< 10	< 10	27	10	112
358006 H	205 238	96	0.02	10	1240	8	5	17	55	0.33	< 10	< 10	255	10	90
358007 H	205 238	17	< 0.01	2	140	6	< 5	2	82	< 0.01	< 10	< 10	37	< 10	12
358008 H	205 238	18	0.02	12	410	10	5	5	126	0.14	< 10	< 10	100	< 10	36
358009 H	205 238	12	0.01	28	1300	12	< 5	6	53	0.24	< 10	< 10	107	< 10	84
358010 H	205 238	22	0.01	39	760	8	5	4	49	0.11	< 10	< 10	100	20	94

CERTIFICATION :

B. Coughlin



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-1C1

PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

Project: CUD89-02

Comments: ATTN: D. CAULFIELD

Page No.: 2-A

Tot. Pages: 2

Date: 9-JUL-89

Invoice #: I-8919181

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8919181

SAMPLE DESCRIPTION	PREP CODE	Au ppb	Al %	Ag ppm	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %	Mn ppm
		FA+AA	%	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	%	ppm	ppm	%	ppm	%	ppm
358011 H	205 238	50	1.73	1.2	15	20	< 0.5	< 2	1.38	< 0.5	143	32	3380	8.65	10	< 1	0.12	< 10	0.53	320
358012 H	205 238	20	1.51	< 0.2	5	20	< 0.5	< 2	2.25	< 0.5	19	51	201	4.11	10	< 1	0.05	< 10	0.68	295
358013 H	205 238	< 5	2.34	< 0.2	< 5	20	< 0.5	< 2	2.69	< 0.5	20	16	189	4.73	10	< 1	0.06	< 10	0.62	425
358014 H	205 238	< 5	2.98	< 0.2	15	30	< 0.5	< 2	4.31	< 0.5	12	35	292	3.00	< 10	< 1	0.10	< 10	0.74	370
358015 H	205 238	< 5	1.98	< 0.2	15	30	< 0.5	< 2	2.27	< 0.5	12	41	302	3.64	10	< 1	0.12	< 10	0.80	265
358016 H	205 238	< 5	2.92	< 0.2	< 5	20	< 0.5	< 2	3.20	< 0.5	19	36	226	3.04	< 10	< 1	0.20	< 10	0.44	180
358017 H	205 238	< 5	2.88	< 0.2	5	20	0.5	< 2	3.09	< 0.5	19	50	384	4.27	10	< 1	0.10	< 10	0.45	200
358018 H	205 238	< 5	3.79	< 0.2	135	< 10	1.5	< 2	6.46	< 0.5	52	38	584	11.25	10	< 1	0.02	< 10	1.16	1270
358019 H	205 238	25	2.02	< 0.2	45	10	1.0	< 2	1.75	< 0.5	21	52	128	6.28	10	< 1	0.04	< 10	0.92	195
358020 H	205 238	80	1.08	< 0.2	30	< 10	1.5	< 2	4.15	< 0.5	81	22	699	>15.00	20	< 1	0.02	< 10	0.36	860
358021 H	205 238	85	2.42	< 0.2	15	10	1.5	< 2	0.83	< 0.5	84	21	1140	>15.00	10	< 1	0.15	10	0.41	270
358022 H	205 238	45	0.65	< 0.2	40	20	< 0.5	< 2	0.08	< 0.5	25	18	311	>15.00	20	< 1	0.16	< 10	0.15	175
358023 H	205 238	< 5	0.17	< 0.2	5	< 10	< 0.5	8	>15.00	< 0.5	2	3	6	0.34	< 10	< 1	0.11	< 10	0.06	160
358024 H	205 238	115	2.28	< 0.2	55	10	0.5	< 2	1.29	< 0.5	29	79	195	6.28	10	1	0.40	10	2.09	250
358025 H	205 238	45	1.52	< 0.2	20	30	0.5	2	0.30	< 0.5	9	76	46	11.85	20	< 1	0.32	10	1.54	320
358026 H	205 238	< 5	0.06	< 0.2	< 5	< 10	< 0.5	8	>15.00	< 0.5	10	3	53	2.11	< 10	< 1	0.01	< 10	0.06	2900
358027 H	205 238	15	1.37	< 0.2	35	< 10	0.5	< 2	14.90	1.0	76	40	884	>15.00	< 10	< 1	0.02	< 10	0.54	940

CERTIFICATION :

B. Caulfield



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Analytical Chemists • Geochemists • Registered Assayers

212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1

PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

Project: CDD89-02

Comments: ATTN: D CAULFIELD

Page No. : 2-B

Tot. Pages: 2

Date : 9-JUL-89

Invoice #: I-8919181

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8919181

SAMPLE DESCRIPTION	PREP CODE		Mo	Na	Ni	P	Pb	Sb	Sc	Sr	Ti	Tl	U	V	W	Zn
			ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm	ppm
358011 H	205	238	22	0.01	13	70	6	< 5	1	133	0.08	< 10	< 10	44	< 10	58
358012 H	205	238	3	0.05	36	1200	6	< 5	5	28	0.25	< 10	< 10	72	< 10	30
358013 H	205	238	2	0.13	9	680	4	< 5	5	41	0.30	< 10	< 10	83	< 10	20
358014 H	205	238	8	0.03	14	970	< 2	< 5	7	87	0.20	< 10	< 10	118	< 10	52
358015 H	205	238	1	0.04	19	1230	< 2	5	5	101	0.22	< 10	< 10	127	< 10	44
358016 H	205	238	1	0.15	22	1060	< 2	< 5	4	63	0.22	< 10	< 10	99	< 10	30
358017 H	205	238	< 1	0.04	26	1120	12	5	4	32	0.23	< 10	< 10	154	< 10	32
358018 H	205	238	1	0.02	58	1130	6	5	10	46	0.12	< 10	< 10	191	< 10	52
358019 H	205	238	4	0.03	30	1230	8	< 5	8	50	0.18	< 10	< 10	125	< 10	24
358020 H	205	238	17	< 0.01	5	460	4	10	3	29	0.09	< 10	< 10	61	< 10	42
358021 H	205	238	4	0.25	31	650	< 2	5	4	59	0.11	< 10	< 10	56	< 10	48
358022 H	205	238	74	< 0.01	10	610	12	10	3	7	0.07	< 10	< 10	49	< 10	280
358023 H	205	238	1	< 0.01	< 1	30	4	5	< 1	725	< 0.01	< 10	< 10	3	< 10	10
358024 H	205	238	2	0.03	38	1270	4	5	7	14	0.28	< 10	< 10	150	< 10	24
358025 H	205	238	53	< 0.01	11	1040	34	5	8	14	0.10	< 10	< 10	131	< 10	124
358026 H	205	238	< 1	< 0.01	< 1	10	114	5	2	1145	< 0.01	< 10	< 10	3	< 10	12
358027 H	205	238	3	0.01	28	430	48	5	4	237	0.07	< 10	< 10	29	< 10	246

CERTIFICATION :

B. Coughlin



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers
 212 BROOKSBANK AVE., NORTH VANCOUVER,
 BRITISH COLUMBIA, CANADA V7J-2C1
 PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
 VANCOUVER, BC
 V6B 1N2

A8919182

Comments: ATTN: D. CAULFIELD

CERTIFICATE A8919182

EQUITY ENGINEERING LTD.

PROJECT : CDD89-02

P O # : NONE

Samples submitted to our lab in Vancouver, BC.
 This report was printed on 9-JUL-89.

SAMPLE PREPARATION

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION
201	10	Dry, sieve -80 mesh; soil, sed.
238	10	ICP: Aqua regia digestion

• NOTE 1:

The 32 element ICP package is suitable for trace metals in soil and rock samples. Elements for which the nitric-aqua regia digestion is possibly incomplete are: Al, Ba, Be, Ca, Cr, Ga, K, La, Mg, Na, Sr, Ti, Tl, W.

ANALYTICAL PROCEDURES

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	UPPER LIMIT
100	10	Au ppb: Fuse 10 g sample	FA-AAS	5	10000
921	10	Al %: 32 element, soil & rock	ICP-AES	0.01	15.00
922	10	Ag ppm: 32 element, soil & rock	ICP-AES	0.2	200
923	10	As ppm: 32 element, soil & rock	ICP-AES	5	10000
924	10	Ba ppm: 32 element, soil & rock	ICP-AES	10	10000
925	10	Be ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
926	10	Bi ppm: 32 element, soil & rock	ICP-AES	2	10000
927	10	Ca %: 32 element, soil & rock	ICP-AES	0.01	15.00
928	10	Cd ppm: 32 element, soil & rock	ICP-AES	0.5	100.0
929	10	Co ppm: 32 element, soil & rock	ICP-AES	1	10000
930	10	Cr ppm: 32 element, soil & rock	ICP-AES	1	10000
931	10	Cu ppm: 32 element, soil & rock	ICP-AES	1	10000
932	10	Fe %: 32 element, soil & rock	ICP-AES	0.01	15.00
933	10	Ga ppm: 32 element, soil & rock	ICP-AES	10	10000
934	10	Hg ppm: 32 element, soil & rock	ICP-AES	1	10000
935	10	K %: 32 element, soil & rock	ICP-AES	0.01	10.00
936	10	La ppm: 32 element, soil & rock	ICP-AES	10	10000
937	10	Mg %: 32 element, soil & rock	ICP-AES	0.01	15.00
938	10	Mn ppm: 32 element, soil & rock	ICP-AES	5	10000
939	10	Mo ppm: 32 element, soil & rock	ICP-AES	1	10000
940	10	Na %: 32 element, soil & rock	ICP-AES	0.01	5.00
941	10	Ni ppm: 32 element, soil & rock	ICP-AES	1	10000
942	10	P ppm: 32 element, soil & rock	ICP-AES	10	10000
943	10	Pb ppm: 32 element, soil & rock	ICP-AES	2	10000
944	10	Sb ppm: 32 element, soil & rock	ICP-AES	5	10000
945	10	Sc ppm: 32 elements, soil & rock	ICP-AES	1	100000
946	10	Sr ppm: 32 element, soil & rock	ICP-AES	1	10000
947	10	Ti %: 32 element, soil & rock	ICP-AES	0.01	5.00
948	10	Tl ppm: 32 element, soil & rock	ICP-AES	10	10000
949	10	U ppm: 32 element, soil & rock	ICP-AES	10	10000
950	10	V ppm: 32 element, soil & rock	ICP-AES	1	10000
951	10	W ppm: 32 element, soil & rock	ICP-AES	10	10000
952	10	Zn ppm: 32 element, soil & rock	ICP-AES	2	10000



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212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1

PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

Project: CDD89-02

Comments: ATTN: D. CAULFIELD

Page No. : 1-A

Tot. Pages: 1

Date : 9-JUL-89

Invoice #: I-8919182

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8919182

SAMPLE DESCRIPTION	PREP CODE		Au ppb	Al %	Ag ppm	As ppm	Ba ppm	Be ppm	Bi ppm	Ca %	Cd ppm	Co ppm	Cr ppm	Cu ppm	Fe %	Ga ppm	Hg ppm	K %	La ppm	Mg %	Mn ppm
			FA+AA																		
DAC S-1	201	238	90	3.26	0.8	20	90	1.0	< 2	2.91	< 0.5	41	89	720	7.38	10	< 1	0.17	< 10	2.64	1680
DAC S-2	201	238	40	3.23	0.2	30	70	1.0	< 2	2.25	< 0.5	38	59	384	6.64	< 10	2	0.13	< 10	2.18	1680
DAC S-3	201	238	100	2.96	0.4	20	150	2.0	< 2	3.82	< 0.5	39	43	595	7.53	< 10	< 1	0.11	< 10	2.42	1345
DAC S-4	201	238	< 5	3.12	< 0.2	15	50	0.5	< 2	1.72	< 0.5	23	79	87	4.83	< 10	< 1	0.06	< 10	1.94	835
DAC S-5	201	238	< 5	1.69	0.2	10	40	1.0	< 2	4.00	< 0.5	28	54	221	9.39	< 10	< 1	0.13	< 10	1.02	1070
DAC S-6	201	238	< 5	2.41	0.2	25	30	1.0	< 2	3.94	< 0.5	17	31	352	5.24	< 10	1	0.10	< 10	1.02	1070
DAC S-7	201	238	< 5	1.78	0.2	15	180	0.5	< 2	4.10	< 0.5	22	32	125	5.37	< 10	1	0.40	< 10	1.31	940
DAC S-8	201	238	45	2.52	0.4	25	60	1.0	< 2	6.24	< 0.5	26	30	257	7.81	< 10	< 1	0.50	< 10	2.08	1765
TB-1	201	238	15	3.22	< 0.2	30	70	1.0	< 2	1.57	< 0.5	29	47	197	5.65	< 10	< 1	0.17	10	1.80	1120
TB-2	201	238	< 5	2.86	< 0.2	20	50	0.5	< 2	1.83	< 0.5	22	39	81	4.61	< 10	1	0.05	< 10	1.60	930

CERTIFICATION :

B. Caulfield



Chemex Labs Ltd.

Analytical Chemists * Geochemists * Registered Assayers

212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1

PHONE (604) 964-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

Project: CLD89-01

Comments: ATTN: D. CAULFIELD

Page No. : 1-B

Tot. Pages: 1

Date : 9-JUL-89

Invoice #: I-8919182

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8919182

SAMPLE DESCRIPTION	PREP CODE		Mo	Na	Ni	P	Pb	Sb	Sc	Sr	Ti	Tl	U	V	W	Zn
			ppm	%	ppm	ppm	ppm	ppm	ppm	ppm	ppm	%	ppm	ppm	ppm	ppm
DAC S-1	201	238	17	0.02	26	1590	10	5	21	179	0.29	< 10	< 10	252	< 10	122
DAC S-2	201	238	4	0.02	21	1000	14	< 5	19	116	0.22	< 10	< 10	209	< 10	114
DAC S-3	201	238	9	0.02	15	2730	4	5	19	403	0.36	< 10	< 10	303	< 10	100
DAC S-4	201	238	1	0.04	41	1200	4	< 5	14	226	0.34	< 10	< 10	169	< 10	104
DAC S-5	201	238	< 1	0.06	15	3840	< 2	5	10	290	0.18	< 10	< 10	380	< 10	90
DAC S-6	201	238	< 1	0.04	12	1590	10	< 5	10	200	0.36	< 10	< 10	282	< 10	80
DAC S-7	201	238	< 1	0.21	10	4970	< 2	5	11	584	0.25	< 10	< 10	239	< 10	76
DAC S-8	201	238	< 1	0.13	9	4460	4	5	23	716	0.51	< 10	< 10	438	< 10	138
TB-1	201	238	< 1	0.05	24	1180	14	< 5	13	137	0.26	< 10	< 10	188	< 10	114
TB-2	201	238	< 1	0.02	20	1130	2	< 5	11	191	0.26	< 10	< 10	142	< 10	90

CERTIFICATION :

B. Coughlin



Chemex Labs Ltd.

Analytical Chemists • Geochemists • Registered Assayers
212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1
PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

A8920703

Comments: ATTN: D. CAULFIELD

CERTIFICATE A8920703

EQUITY ENGINEERING LTD
PROJECT : CDD89-02
P. O. # : NONE

Samples submitted to our lab in Vancouver, BC.
This report was printed on 17-JUL-89.

SAMPLE PREPARATION

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION
214	1	Received sample as pulp

ANALYTICAL PROCEDURES

CHEMEX CODE	NUMBER SAMPLES	DESCRIPTION	METHOD	DETECTION LIMIT	UPPER LIMIT
396	1	Au oz/T: 1/2 assay ton	FA-GRAVIMETRIC	0.003	20.000



Chemex Labs Ltd.

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212 BROOKSBANK AVE., NORTH VANCOUVER,
BRITISH COLUMBIA, CANADA V7J-2C1

PHONE (604) 984-0221

To: EQUITY ENGINEERING LTD.

207 - 675 W. HASTINGS ST.
VANCOUVER, BC
V6B 1N2

Project: CDD89-02

Comments: ATTN: D. CAULFIELD

Page No.: 1

Tot. Pages: 1

Date: 17-JUL-89

Invoice #: I-8920703

P.O. #: NONE

CERTIFICATE OF ANALYSIS A8920703

SAMPLE DESCRIPTION	PREP CODE	Au FA oz/T										
30268 E	214	--	0.422									

CERTIFICATION :

W. Santamaria
B.C. CERTIFIED ASSAYER

APPENDIX E

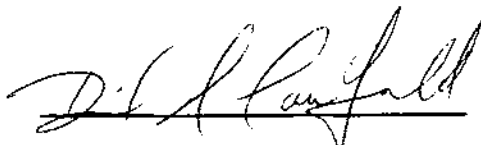
STATEMENT OF QUALIFICATIONS

STATEMENT OF QUALIFICATIONS

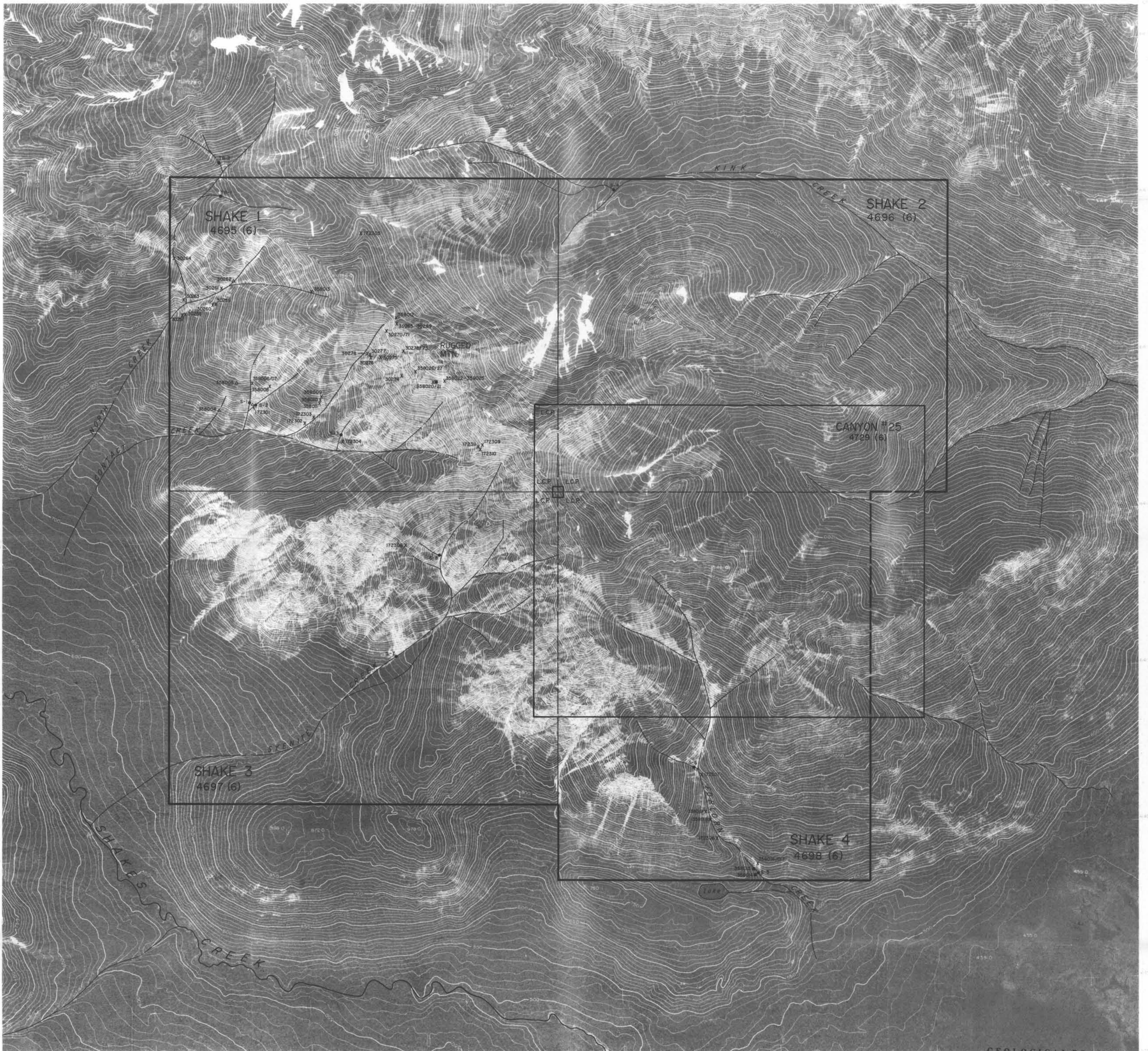
I, DAVID A. CAULFIELD, of 3142 Gambier Street, Coquitlam, in the Province of British Columbia, DO HEREBY CERTIFY:

1. THAT I am a Consulting Geologist with offices at Suite 207, 675 West Hastings Street, Vancouver, British Columbia.
2. THAT I am a graduate of the University of British Columbia with a Bachelor of Science degree in Geology.
3. THAT my primary employment since 1978 has been in the field of mineral exploration.
4. THAT my experience has encompassed a wide range of geological environments and has allowed considerable familiarization with geophysical, geochemical, and diamond drilling techniques.
5. THAT I am a Fellow, in good standing of the Geological Association of Canada.
6. THAT this report is based on fieldwork carried out under my direct supervision from June 16 through June 23, 1989 and on government publications and assessment reports filed with the Province of British Columbia.

DATED at Vancouver, British Columbia, this 25 day of September, 1989.



David A. Caulfield, F.G.A.C.



GEOLOGICAL BRANCH
ASSESSMENT REPORT

19,127
SCALE 1:10,000

m 0 200 400 600 800 1000 m

ROCK GEOCHEMICAL RESULTS

Sample	As(ppb)	As(ppm)	Cu(ppm)	Pb(ppm)	Zn(ppm)	As(ppm)
30257	15	1.0	814	<2	42	40
30259	35	1.2	1220	12	42	55
30259	20	0.6	523	8	62	60
30260	40	<0.2	280	2	16	25
30261	60	0.6	924	<2	114	25
30262	15	0.2	134	2	26	10
30263	20	1.4	342	<2	62	30
30264	25	1.2	409	10	66	15
30265	730	0.4	80	14	48	>10000
30266	5370	2.0	178	24	66	8500
30267	370	<0.2	54	6	32	3120
30268	6.8	80	14	40	5300	
30269	200	1.4	993	2	50	155
30270	375	0.2	23	32	20	1495
30271	180	<0.2	15	48	106	1215
30272	20	0.8	23	468	720	40
30273	40	0.8	133	42	54	30
30274	35	0.6	725	42	82	60
30275	680	1.8	328	6	40	35
30276	50	0.2	318	14	188	40
30277	35	<0.2	79	6	986	25
30278	55	<0.2	136	14	106	15
172301	25	<0.2	855	42	72	5
172302	790	59.0	>10000	10	522	60
172303	45	1.2	1095	42	114	30
172304	45	<0.2	533	18	66	10
172305	5	<0.2	89	18	80	10
172306	5	<0.2	165	42	84	30
172307	110	9.8	3980	24	124	10
172308	1120	2.0	127	28	32	20
172309	95	3.2	1010	12	42	25
172310	60	1.4	242	8	20	10
172311	120	2.8	477	42	28	20
358004	30	<0.2	319	28	66	5
358005	<5	1.2	484	<2	112	<5
358006	<5	0.6	1725	8	90	5
358007	260	<0.2	44	6	12	35

* 0.422 oz/ton Au

SILT GEOCHEMICAL RESULTS

Sample	As(ppb)	As(ppm)	Cu(ppm)	Pb(ppm)	Zn(ppm)	As(ppm)
DAC S-1	90	0.8	720	10	122	20
DAC S-2	40	0.2	384	14	214	30
DAC S-3	100	0.4	595	4	100	20
DAC S-4	45	<0.2	87	4	104	15
DAC S-5	<5	0.2	221	<2	90	10
DAC S-6	<5	0.2	352	10	80	25
DAC S-7	<5	0.2	125	<2	76	15
DAC S-8	45	0.4	207	4	138	25
TB-1	15	<0.2	197	14	114	30
TB-2	<5	<0.2	81	2	90	20

LEGEND

X Δ ■ ROCK SAMPLE (GRAB-OUTCROP, FLOAT, CHIP)

● SILT SAMPLE

CANDELA RESOURCES LTD.

SHAKE 1-4 CLAIM GROUP

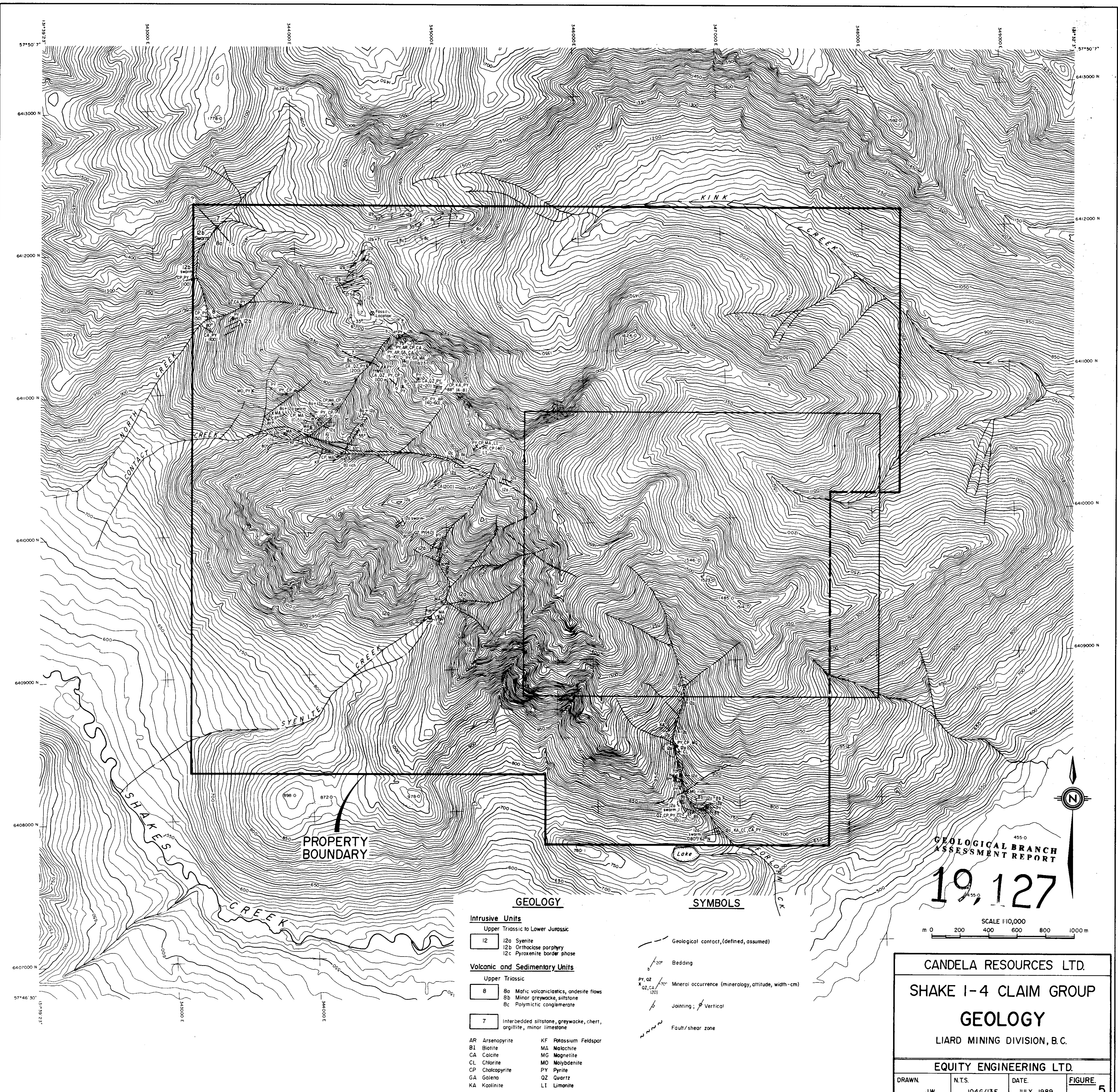
ROCK & STREAM

GEOCHEMISTRY

LIARD MINING DIVISION, B.C.

EQUITY ENGINEERING LTD.

DRAWN.	N.T.S.	DATE.	FIGURE.
J.W.	104 G/13E	JULY, 1989	4



GEOLOGICAL BRANCH
ASSESSMENT REPORT
19,127

SCALE 1:10,000
m 0 200 400 600 800 1000 m

GEOLOGY

- Intrusive Units**
Upper Triassic to Lower Jurassic
- 12 12a Syenite
 - 12b Orthoclase porphyry
 - 12c Pyroxenite border phase
- Volcanic and Sedimentary Units**
Upper Triassic
- 8 8a Mafic volcanics, andesite flows
 - 8b Minor greywacke, siltstone
 - 8c Polymictic conglomerate
 - 7 Interbedded siltstone, greywacke, chert, argillite, minor limestone
- | | |
|-----------------|-----------------------|
| AR Arsenopyrite | KF Potassium Feldspar |
| B1 Biotite | MA Malachite |
| CA Calcite | MG Magnetite |
| CL Chlorite | MO Molybdenite |
| CP Chalcopyrite | PY Pyrite |
| GA Galena | QZ Quartz |
| KA Kaolinite | LI Limonite |

SYMBOLS

- Geological contact, (defined, assumed)
- Bedding
- Mineral occurrence (mineralogy, attitude, width-cm)
- Jointing; Vertical
- Fault/shear zone

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GEOLOGY			
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EQUITY ENGINEERING LTD.			
DRAWN J.W.	N.T.S. 1046/13E	DATE JULY, 1989	FIGURE 5