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GEOLOGICAL and GEOCHEMICAL

REPORT

on the

ABE PROPERTY

Aiken Lake area

Omineca Mining District

British Columbia

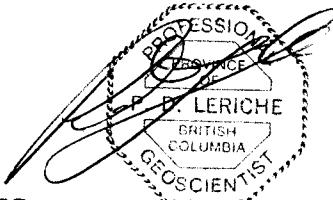
**GEOLOGICAL BRANCH
 ASSESSMENT REPORT**

22,860

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23 February 1993

SUMMARY

At the request of the Swannell Minerals Corporation, Reliance Geological Services carried out an exploration program consisting of rock and stream sediment surveys and geological mapping on the ABE property during July and August 1992.

The ABE property comprises twelve contiguous mineral claims totalling 140 units in the Aiken Lake area of the Omineca Mining Division. The property is situated approximately 225 kilometers northwest of Fort St James, B.C., and is accessible by helicopter.

The claims lie in the regionally extensive Mesozoic Quesnel Belt where alkalic plutons commonly host porphyry copper-gold deposits. In the Aiken Lake district, Triassic Takla volcanic rocks are intruded by Triassic-Jurassic alkaline stocks and Cretaceous calk-alkaline stocks of the Hogem Batholith.

The claims are underlain by Triassic-Jurassic Takla volcanics which are intruded by pyroxenite, diorite, monzonite and quartz monzonite of the Triassic-Jurassic Hogem batholith.

Previous work consisted of regional aeromagnetic, soil, and silt sampling surveys, and one short drill hole. Four magnetic highs and one 550 by 1550 meter molybdenum were defined.

In 1991, Swannell contracted a silt sampling, rock sampling, and reconnaissance mapping program. Copper and gold mineralization was encountered in two areas and found to consist of:

- a) flat-lying quartz veins carrying sporadic pyrite-chalcopyrite-galena-hematite,
- b) fracture-controlled malachite-chalcopyrite-magnetite, and
- c) disseminated chalcopyrite-pyrite associated with strong carbonate-chlorite alteration.

The 1992 program identified two target areas:

First: centered around an outcrop which consists of fracture-filled and disseminated chalcopyrite in a chlorite/sericite/K-feldspar altered porphyritic andesite.

A one meter chip sample assayed 2202 ppm Cu and 117 ppb Au. The isolated outcrop is in a talus filled valley, and the general area has yet to be investigated.

Second: an open-ended, 400 by 600 meter copper soil anomaly with anomalous gold results forming a peripheral pattern.

Further work consisting of geological mapping, and magnetic and IP geophysics has been recommended to establish drill targets. Contingent on favorable results, diamond drilling is recommended to test the targets at depth.

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1.

INTRODUCTION

This report was prepared at the request of Swannell Minerals Corporation to describe and evaluate the results of the 1992 geological and geochemical program carried out by Reliance Geological Services Inc on the ABE claim group in the Johanson Lake area of the Omineca Mining District, British Columbia.

The field work was undertaken for the purpose of following up on the 1991 program and evaluating the potential of the property to host porphyry copper/gold deposits.

Field work was carried out from July 25 to 27, 1992, by Alan Taylor (project geologist), Doug Johannessen (geologist), Brian Chure (geo-technician), and Ted Archibald (prospector). A follow-up program was carried out on August 22, 23, and 26, 1992, by Alan Taylor (project geologist), Reg Faulkner (geologist), George King (geologist), and John Fleishman (prospector). All work was carried out under the supervision of Peter Leriche, P.Geo.

This report is based on published and unpublished information and the maps, reports and field notes of the crew listed above.

2. LOCATION, ACCESS and PHYSIOGRAPHY

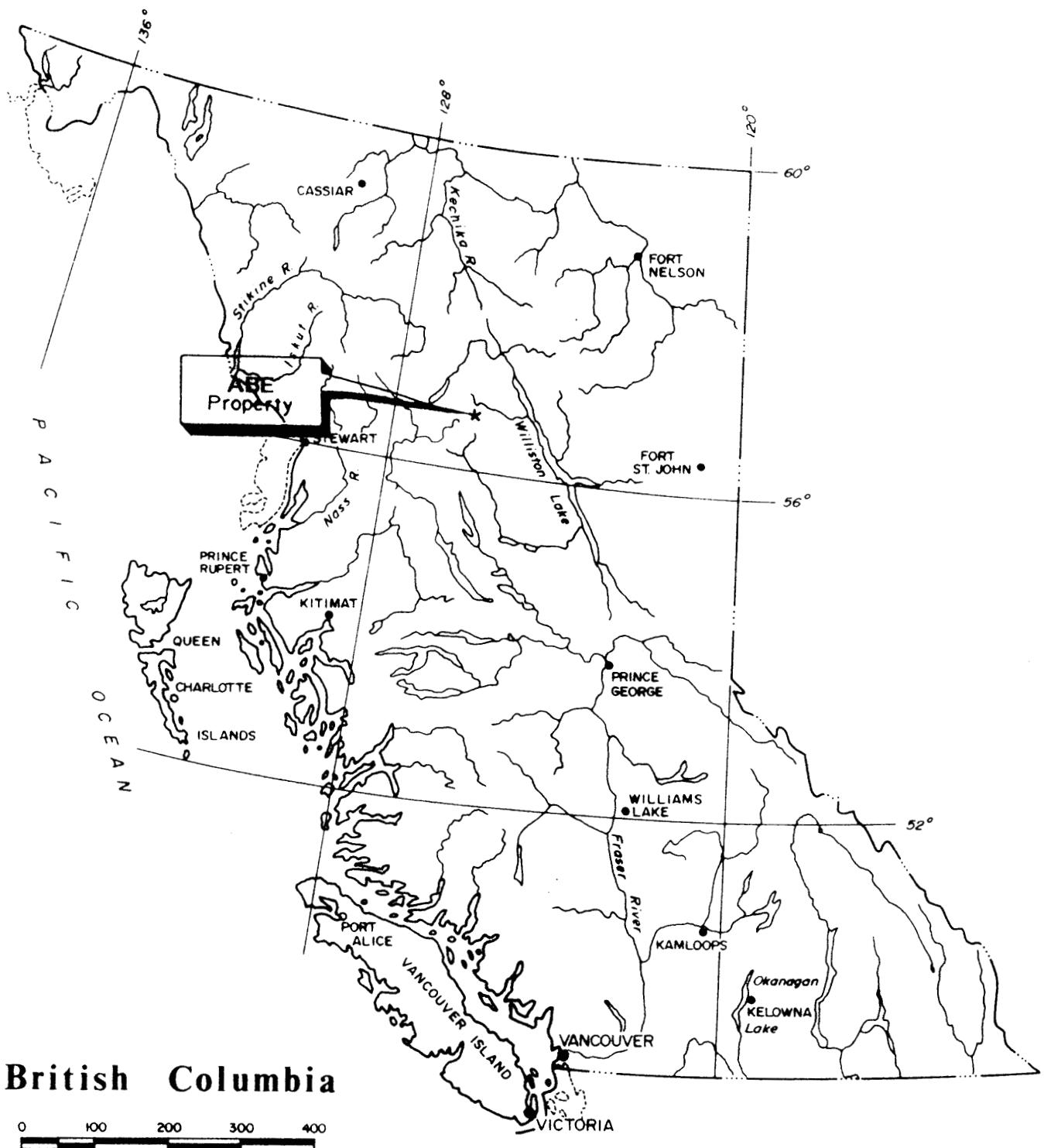
The ABE property is situated in the Omineca Mining Division in the Aiken Lake area, approximately 235 kilometers northwest of Fort St James (Figures 1 and 2).

The claims are located on Map Sheet NTS 94C/5E & 5W, at latitude 56° 21' North, longitude 125° 48' West, and between UTM 6251000 m and 6245500 m North, and UTM 325000 m and 333000 m East.

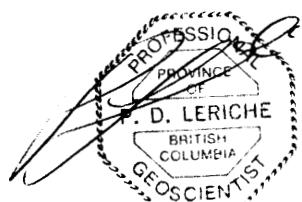
Road access is via the Omineca Mining Road from Fort St James north to Aiken Lake (approximately 360 km), then by helicopter from the seasonal base at Aiken Lake.

The property is on mountainous terrain with moderate to steep slopes rising from approximately 1320 meters to 2000 meters. The area is sparsely forested with spruce and pine. Scrub fir and alpine vegetation occur above tree-line (\pm 1600 meters).

Recommended work season is mid-June to early October.



British Columbia



SWANNELL MINERALS CORPORATION

ABE PROPERTY

General Location Map

Scale noted above	N.T.S.	Drawn by
Date JAN. 93	Geologist	Figure 1
RELIANCE GEOLOGICAL SERVICES INC.		

3. PROPERTY STATUS

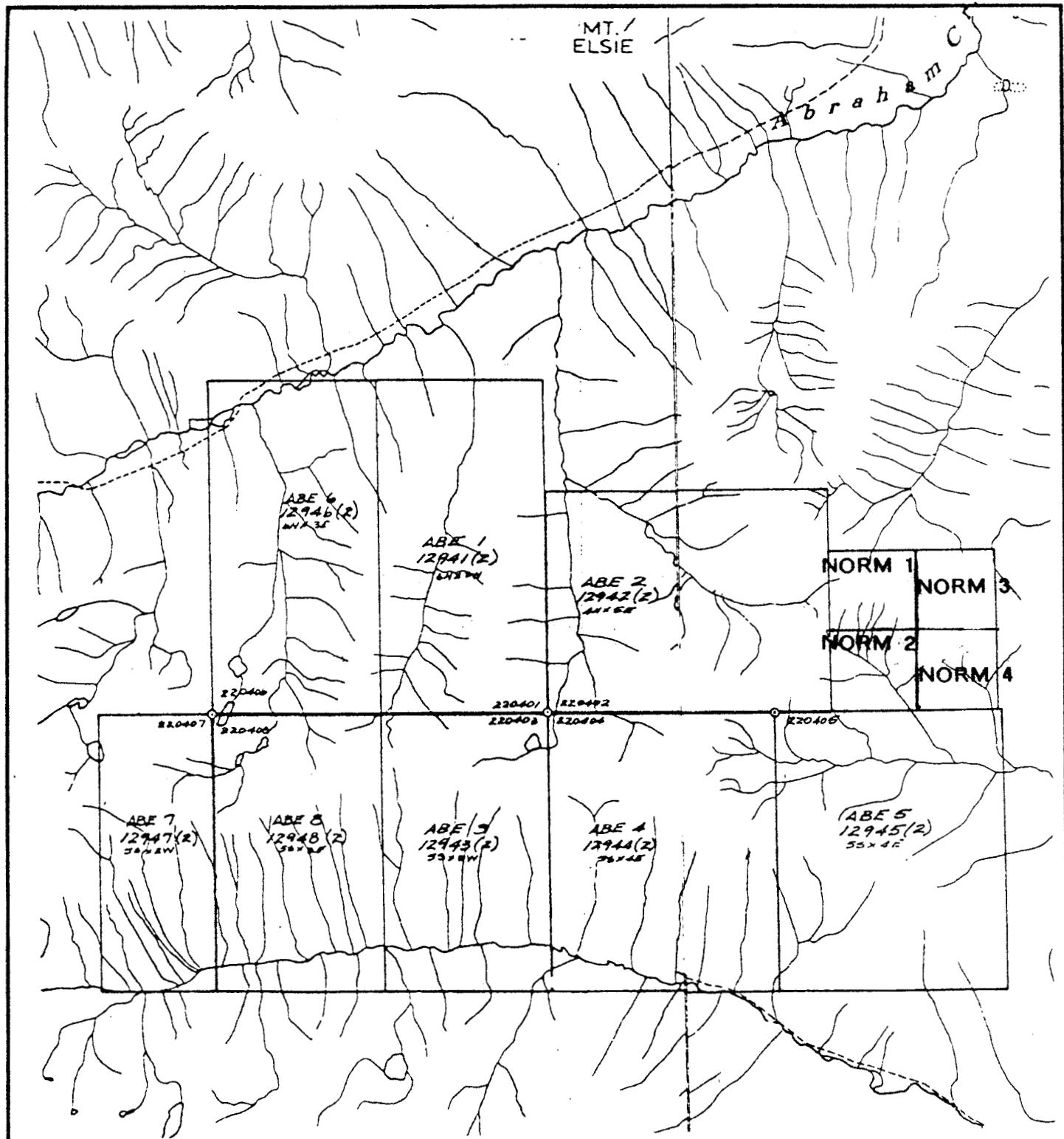
The property consists of twelve contiguous mineral claims (Figure 2) in the Omineca Mining Division. The claims are registered in the name of Major General Resources Ltd and Harvey Keck, and have been optioned to Swannell Minerals Corporation.

Details of the claims are as follows:

<u>Claim</u>	<u>Record Number</u>	<u>Units</u>	<u>Record Date</u>	<u>Expiry Date</u>
Abe 1	243091	18	9 Feb 1991	9 Feb 1995
Abe 2	243092	20	9 Feb 1991	9 Feb 1995
Abe 3	243093	15	10 Feb 1991	10 Feb 1995
Abe 4	243094	20	10 Feb 1991	10 Feb 1995
Abe 5	243095	20	9 Feb 1991	9 Feb 1995
Abe 6	243096	18	9 Feb 1991	9 Feb 1995
Abe 7	243097	10	10 Feb 1991	10 Feb 1995
Abe 8	243098	15	10 Feb 1991	10 Feb 1995
Norm 1	312957	1	22 Aug 1992	22 Aug 1995
Norm 2	312958	1	22 Aug 1992	22 Aug 1995
Norm 3	312959	1	22 Aug 1992	22 Aug 1995
Norm 4	312960	1	22 Aug 1992	22 Aug 1995
Total		140 units		

The total area covered by the claims is 3500 hectares, or 8645 acres, allowing for overlap.

The writers are not aware of any particular environmental, political or regulatory problems that would adversely affect mineral exploration and development on the ABE property.



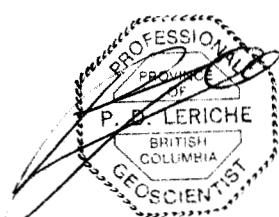
SWANNELL MINERALS CORPORATION

ABE PROPERTY

CLAIM MAP

Scale 1:50,000 N.T.S. 94C/5W,5E Drawn by
Date JAN. 93 Geologist Figure 2

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4.

REGIONAL GEOLOGY

(from Rebagliati, 1991)

"The ABE property lies within the regionally extensive early Mesozoic Quesnel Belt. This 35 km wide belt extends northwesterly for 1200 km and includes equivalent rocks of the Upper Triassic-Lower Jurassic Takla, Nicola, and Stuhini Groups (Mortimer, 1986) (Figures 3 and 4). To the west, deformed and uplifted Permian Cache Creek Group rocks are separated from the Quesnel Belt by the Pinchi Fault Zone. To the east, the Manson Fault Zone separates this belt from the uplifted Proterozoic/early Palaeozoic Wolverine Metamorphic Complex, and the Mississippian-Permian Slide Mountain and Cache Creek Groups (Garnet, 1978).

In the Mt. Milligan - Johanson Lake district, the Takla Group volcanics are dominated by subaqueous alkalic to subalkalic dark green tuffs and volcanic breccias of andesitic and basaltic composition, interbedded with pyroxene porphyritic flow rocks of similar composition. Intercalated bedded tuffs and argillites are subordinate. Black argillites interfinger with volcanic rocks to the east and west of the central volcanic core. Locally, thick successions of maroon coloured lahars suggest the presence of emergent subaerial volcanic centres.

The volcanic-sedimentary strata of the Quesnel Belt are locally intruded by alkaline syenite, monzonite, and diorite batholiths, stocks and dykes. In the Quesnel Belt, most intrusions are considered coeval and comagmatic with late Triassic-early Jurassic volcanism. Many of the stocks lie along linear trends which are interpreted to reflect fault zones which have localized volcanism and associated stock emplacement.

The Hogem Batholith of Early Jurassic to Cretaceous age is the largest body of intrusive rock within the Omineca Mountains (Armstrong and Garnett 1973) (Figure 4). Takla Group volcanic and sedimentary strata are intruded by the north-south elongate batholith which is, in part, truncated along its western margin by the Pinchi Fault. Numerous satolithic plutons flank the eastern margins of the batholith.

The complexity of the Hogem Batholith is characterized by rock units ranging in composition from diorite to granite. Lithologic changes are rapid to gradational at all scales of mapping.

Garnett, who used the I.U.G.S. classification of 1973 as shown in Table 1 on the following page, described three phases within the Hogem Batholith.

The earliest, Phase I, contains the more basic phases, including pyroxenite, gabbro, diorite, monzodiorite, monzonite, and the "Hogem Granodiorite", and accounts for two-thirds of all rock types mapped. The Hogem Granodiorite is a distinctive leucocratic felsic division, predominantly quartz diorite in composition, but also comprising quartz monzodiorite, quartz monzonite and, more rarely, quartz diorite, tonalite and granite.

The Phase II syenites, such as the Duckling Creek complex, (with migmatitic, compositionally banded, and intrusive varieties) and the leucocratic Chuchi (quartz) syenite, are reported to be intrusive into Phase I rocks.

Phase III rocks include leucocratic varieties (including aplites, pegmatite, varieties of granite, quartz syenite and alaskite). These rocks may be represented by leucocratic late-stage dykes cutting units of Phases I and II.

Numerous porphyry copper prospects occur throughout the Hogem Batholith.

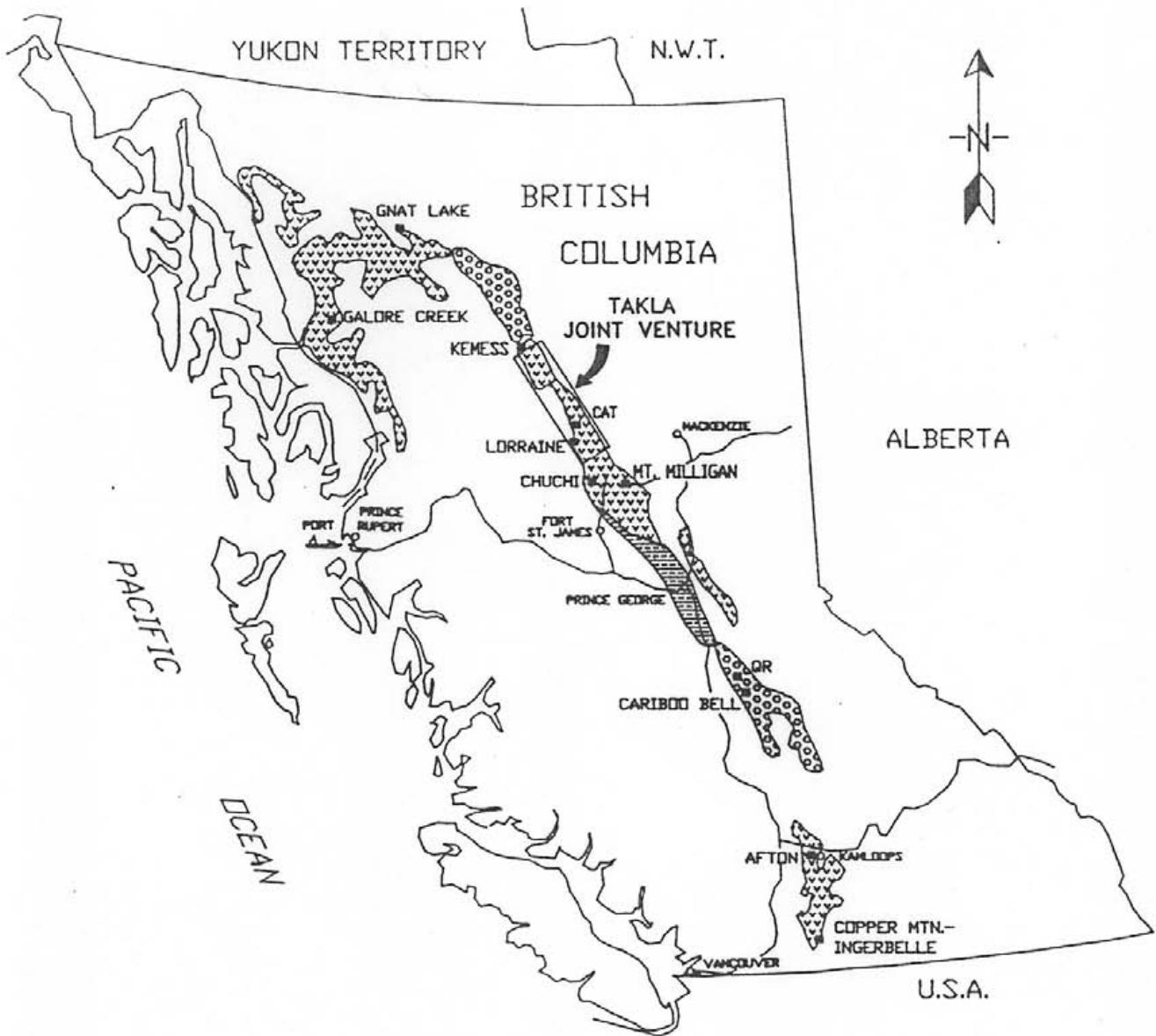
The alkalic plutons of the Quesnel Belt commonly host porphyry copper deposits, which are increasingly being recognized as an important source of gold. It has also been recently recognized that related failed porphyry systems (those that did not form copper deposits) also have the potential to generate disseminated gold deposits (eg: QR and the 66 Zone at Mt Milligan).

The volcanic strata on all of the ABE property claims are intruded by alkalic plutons. Some of these plutons are reported to display some of the geological characteristics which are related to the formation of gold-rich porphyry copper deposits in the Quesnel Belt."

Many auriferous porphyry copper prospects are under active exploration within the Quesnel Belt, and the following deposits have been identified:

Gold-Copper Porphyry Deposits
Quesnel Belt
British Columbia

<u>Property</u>	<u>No. of Deposits</u>	<u>Reserves/Mineral Inventory</u>	
		Copper(x10 ⁶ lbs)	Gold (x10 ⁶ oz)
<u>In Production:</u>			
Copper Mountain (Cassiar)	5	1,600	.910
Afton (Teck)	2	680	.970
<u>Exploration/Development Stage</u>			
Mt Polley (Imperial Metals)	2	875	2.000
Galore Creek (Hudsons Bay et al)	8	3,000	1.750
Red Chris (Noranda)	2	550	.450
QR (QPX)	4	-0-	.200
Lorraine (Kennco)	2	150	.100
Mt. Milligan (Continental Gold/Placer Dome)	2	1,680	6.376
Kemess (El Condor)	2	1,615	6.226



LEGEND

- [Alkaline Volcanic Rocks Pattern] ALKALINE VOLCANIC ROCKS
- [Subalkaline Volcanic Rocks Pattern] SUBALKALINE VOLCANIC ROCKS
- [Mainly Sedimentary Rocks Pattern] MAINLY SEDIMENTARY ROCKS
- GOLD AND / OR COPPER DEPOSIT



After Fox et. al. 1976

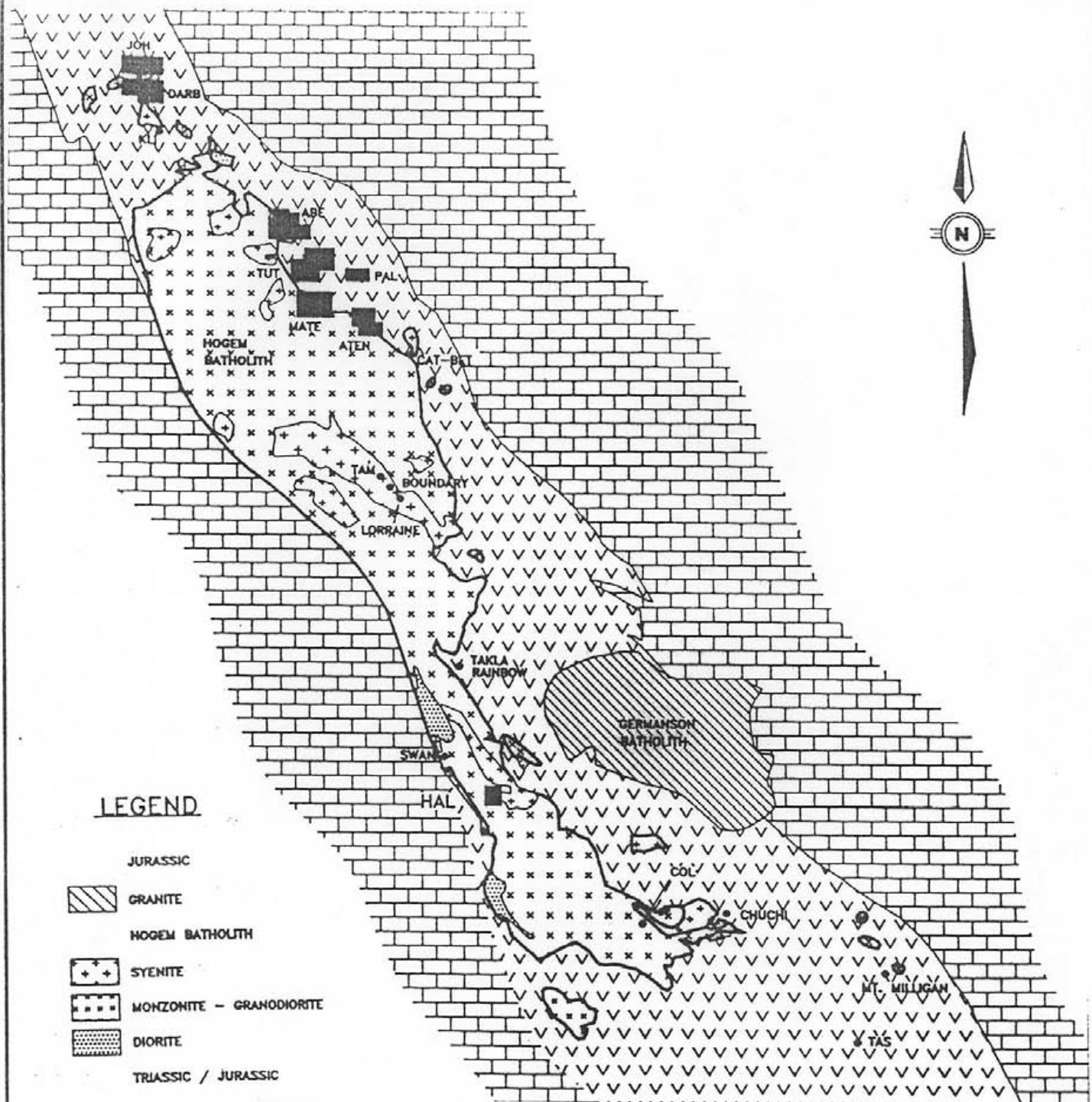
SWANNELL MINERALS CORPORATION

ABE PROPERTY

QUESNEL BELT
UPPER TRIASSIC & LOWER JURASSIC VOLCANIC
ROCKS, SIGNIFICANT GOLD AND / OR COPPER
DEPOSITS, ASSOCIATED WITH ALKALIC PLUTONS

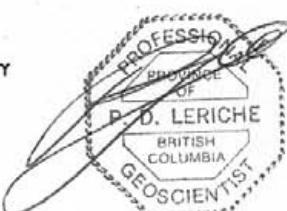
Scale AS SHOWN	N.T.S.	Drawn by
Date JAN. 93	Geologist	Figure 3

6A RELIANCE GEOLOGICAL SERVICES INC.



0 25 50
KILOMETRES

6B



SWANNELL MINERALS CORPORATION

ABE PROPERTY

REGIONAL GEOLOGY

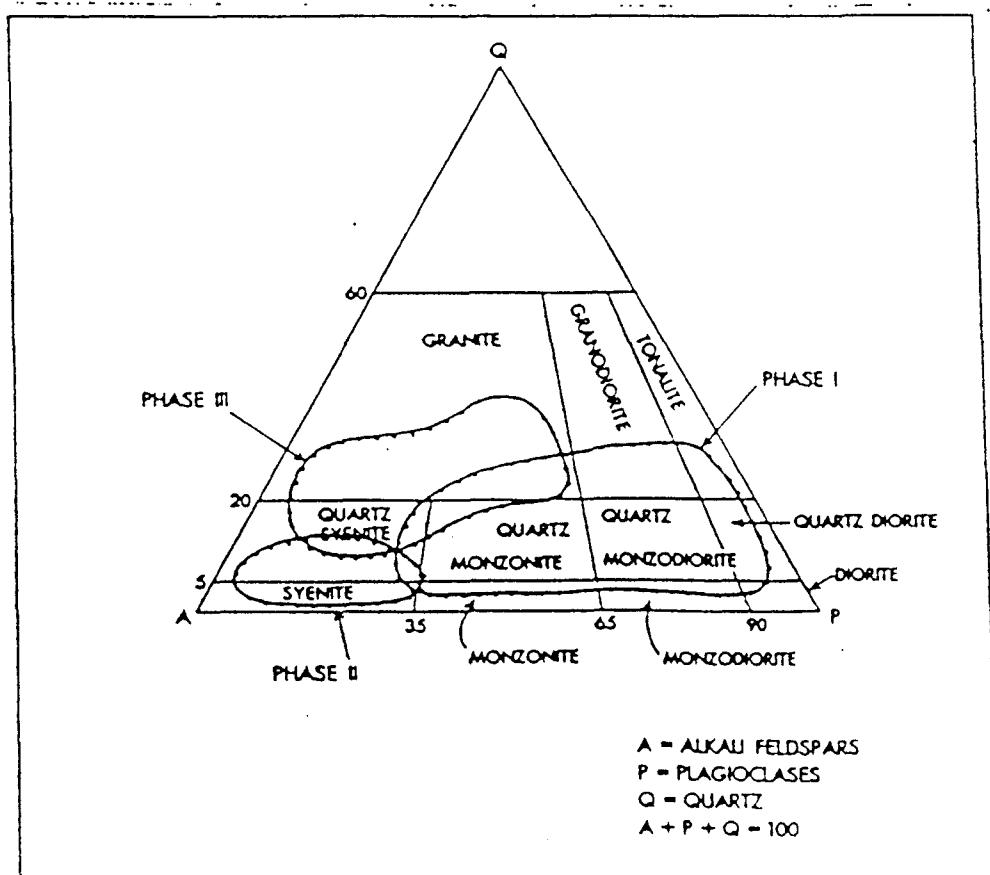
Scale as shown	N.T.S.	Drawn by
Date JAN. 93	Geologist	Figure 4

RELIANCE GEOLOGICAL SERVICES INC.

TABLE 1

SOUTHERN HOGEM BATHOLITH: INTRUSIVE ROCK DIVISIONS

INTRUSIVE PHASES	PHASE DIVISIONS	UNIT	ROCK VARIETIES
PHASE III LOWER CRETACEOUS		9	LEUCOCRATIC GRANITE, Alaskite
PHASE II MIDDLE JURASSIC TO LOWER JURASSIC	CHUCHI SYENITE	8	LEUCOCRATIC SYENITE, Quartz Syenite
	DUCKLING CREEK SYENITE COMPLEX	7	LEUCOCRATIC SYENITE
		6	FOLIATED SYENITE
PHASE I LOWER JURASSIC TO UPPER TRIASSIC	HOGEM GRANODIORITE HOGEM BASIC SUITE	5	GRANODIORITE, QUARTZ MONZONITE, minor Tonalite, Quartz Diorite, Quartz Monzonite, Granite
		4	MONZONITE to Quartz Monzonite
		3	MONZODIORITE to Quartz Monzodiorite
		2	NATION LAKES PLAGIOCLASE PORPHYRY (a) Monzonite (b) Monzodiorite
		1	DIORITE, minor Gabbro, Pyroxenite, Hornblende



Hogem batholith intrusive phases in relation to general plutonic rock classification
After I.U.G.S., 1973.

5. PREVIOUS WORK

Early work included one short hole diamond drilled on the ABE 4 to test a narrow quartz vein mineralized with pyrite, chalcopyrite, galena and tetrahedrite. This vein had been first sampled during the 1950's by the Geological Survey of Canada during regional geologic mapping.

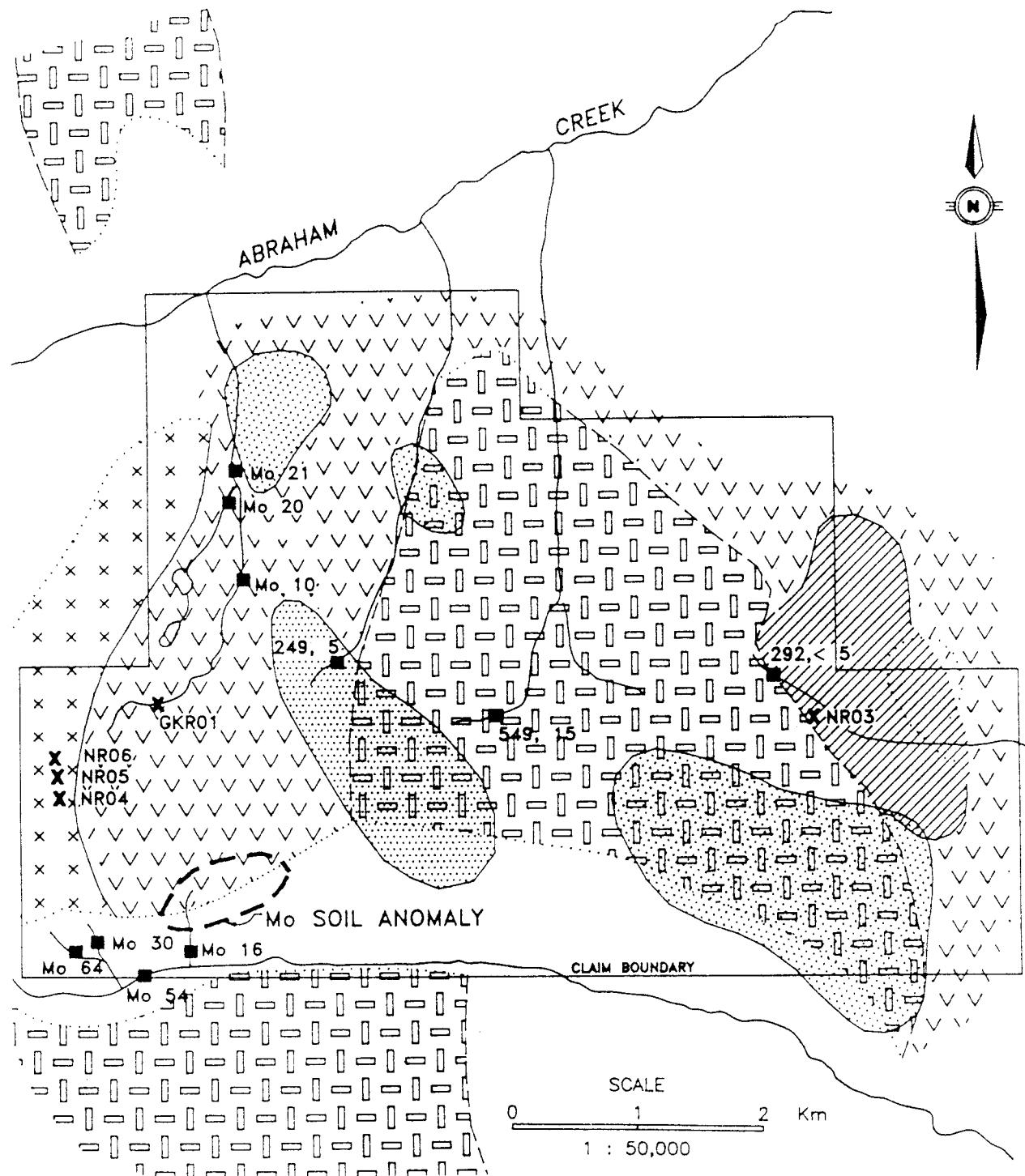
During the 1970's, the ABE claim area was explored by the UMEX-Wenner Gren Joint Venture. The property was covered by regional aeromagnetic, soil and stream sampling surveys.

Four magnetic anomalies were defined (Figure 5): three on the margin of a diorite stock, and one in Takla volcanics near a contact with the Hogem Batholith.

Soil sampling outlined a 550 by 1550 meter molybdenum anomaly on the ABE 7 and 8 claims.

Copper-molybdenum silt anomalies were discovered on three of the streams that drain the property. No samples were analyzed for gold.

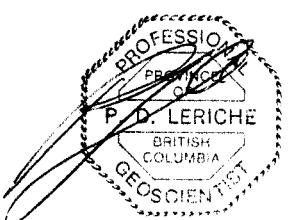
In 1991, Swannell contracted Reliance Geological Services to carry out a program of silt sampling, rock sampling, and 1:10,000 reconnaissance geological mapping. Copper mineralization with values up to 1.28% was found in two areas, one of which had anomalous gold values up to 0.365 oz/t. Anomalous copper and gold values were identified in silts from two streams.



LEGEND

- [Dashed Pattern] DIORITE
- [Diamond Pattern] MONZONITE
- [Hatched Pattern] MAFIC INTRUSION
- [Cross Pattern] HOGEM BATHOLITH (UNDIFF.)
- [V Pattern] TAKLA VOLCANICS
- [Dotted Pattern] AEROMAGNETIC ANOMALY

- SILT ANOMALY ppm Cu, ppb Au, ppm Mo
- HEAVY MINERAL ANOMALY ppm Cu, ppb Au
- ✖ Cu/Mo COPPER / MOLYBDENUM OCCURRENCE



SWANNELL MINERALS CORPORATION		
ABE PROPERTY		
COMPOSITE PLAN		
AEROMAGNETICS GEOLGY AND		
ANOMALOUS STREAM SITES		
Scale 1 : 50,000	N.T.S. 94C/5	Drawn by
Date JAN. 93	Geologist	Figure 5
RELIANCE GEOLOGICAL SERVICES INC.		

6. 1992 WORK PROGRAM

Done under B.C.M.E.M.P.R. Approval Number PRG-1300199-44752

6.1 Methods and Procedures, General

Geological and geochemical surveys were carried out on the claims to follow up on anomalous rock and soil geochemistry identified in previous exploration programs.

A survey grid was laid out over the eastern area of the property. The baseline and tie-line were surveyed using compass, hipchain, and flagging.

Cross-lines were put in at 200 meter line spacings using compass, hipchain, flagging, pickets and metal tags. Stations on baselines and cross-lines were marked at 50 meter intervals using flagging and marked double flagging. Total line surveyed was 21.3 kilometers.

Eleven line kilometers in the northeastern grid area (Lines 98+00N to 108+00N) were cut and marked with embossed metal tags on wooden pickets at 50 meter intervals to prepare for magnetic and induced polarization surveys.

Geological mapping was performed over the grid area at a scale of 1:10,000 (Figure 6).

Thirty-five rock samples were collected and analyzed for gold (FA/AA) and multi-element ICP by International Plasma Laboratory Ltd of Vancouver, B.C.

See Appendix A for rock sample descriptions and Appendix B for analytical reports and techniques.

A polished thin section and a petrographic description of one rock sample was completed by Vancouver Petrographics Ltd, Fort Langley, B.C. (Appendix D)

The current grid was soil sampled at 100 meter station spacings. Using a grub hoe, 158 samples were taken from the B horizon (approximate depth 30 cm), placed into marked Kraft paper bags, and sent to International Plasma Laboratory for analysis.

Forty-one additional soil samples were collected from lines 98+00N, 100+00N, 102+00N, 104+00N, 106+00N, and 108+00N. Samples have not been analyzed.

The analytical results for 2 elements (Cu, Au) were computer-plotted on 1:10,000 scale maps (Figures 7 and 8).

To evaluate any existing geochemical anomalies, frequency distribution histograms based on laboratory data were prepared for each of the aforementioned elements (Appendix C). Anomalous values were chosen using natural breaks in each histogram.

For interpretation purposes, correlation coefficients were calculated (Appendix C) and anomalous ranges for each element were plotted using symbol maps (Figures 7 and 8). All statistical and plotting work was performed by Tony Clark, Ph.D.

6.2 Property Geology (Figure 6)

6.2.1 Lithologies

The ABE grid area is underlain by Triassic - Jurassic Takla volcanics which are intruded by pyroxenite, diorite, monzonite, and quartz monzonite of the Triassic - Jurassic Hogem batholith.

Approximately 40% of the claim area was mapped, with exposed rock along ridges and steeper slopes accounted for approximately 50% of the surface area.

Takla Group:

Unit 1A is an andesite augite porphyry which is exposed on the ridge east of the ABE grid. Augite porphyry commonly weathers a dark brown and, locally where cut by ankeritic dykes, a bright orangy brown. Mineral constituents include 20% (range 10% to 40%) subhedral to euhedral dark green augite phenocrysts in a fine grained matrix of plagioclase and biotite.

Unit 1B consists of fine grained, light grey to dark green andesitic tuffs found in the west central part of the property adjacent to Unit 4B. Tuffs weather to a dark green to rusty brown color. Ankeritic zones with disseminated 1% pyrite are common.

Intrusive Rocks (Hogem Batholith):

Units 2A, 2B, 2C, and 2D are moderately to strongly magnetic green-brown pyroxenites occurring throughout the southern area of the grid.

Unit 2 is variable in texture and has been classified into four local subdivisions:

Unit 2A: fine to medium grained dark green massive pyroxenite.

Unit 2B: sheared, dark black, fine grained chlorite rich phase.

Unit 2C: local coarse gabbroic phase with plagioclase and pyroxene crystals up to 10 cm in length.

Unit 2D: weathers light grey and contains numerous carbonate veins with minor talc and chrysotile.

Unit 3A is a brown-grey, fine to medium grained diorite which is exposed in isolated outcrops on the grid. Minor chlorite and epidote give the rock a light green color. Diorite occurs both as massive outcrop and as dyke-like bodies.

Unit 3B is a hybrid zone located near the periphery of the pyroxenite. It consists of 10% to 90% subangular to angular xenoliths of pyroxenite set in a felsic matrix. The matrix varies from a very fine grained felsite to diorite in composition and weathers to a white chalky color. This zone is distinctive but possibly correlates with Unit 3A.

Unit 3C is a feldspar porphyry which occurs as sub-vertical dyke-like bodies up to 30 meters wide crosscutting pyroxenite. Due to high silica content, Unit 3C forms prominent ridges and pinnacles. Dykes consist of white, subhedral to euhedral, medium grained plagioclase phenocrysts set in a light grey, fine grained silicic matrix containing minor black amphiboles and biotite.

Unit 4A (monzonite) and Unit 4B (quartz monzonite) are found in the western area of the claim. Unit 4 is a massive, coarsely jointed unit. It weathers to a pinkish-brown color, and consists of both medium grained potassium and plagioclase feldspar phenocrysts set in a biotite rich interstitial matrix.

6.2.2 Alteration

Pyroxenite units exhibit local alteration consisting of carbonate-talc veinlets, chloritization and minor serpentine development. Rusty brown weathering quartz-ankerite intrusion-type and replacement-type veins are common within shear zones.

Diorite Unit 3A contains minor epidote-chlorite and exhibits weak pervasive propylitic alteration.

A small outcrop of Takla volcanics in a stream cut north of line 108+00N shows strong chlorite-carbonate alteration.

6.2.3 Structure

Prominent, generally flat-lying sequences of quartz veins cut all local stratigraphy within the pyroxenite units. Quartz crystals within the veins are commonly in a comblike intergrowth indicating static emplacement.

Numerous dykes and shear zones are located in the northern part of the ABE grid. A fault in the northeastern corner of the grid places andesite tuff stratigraphy adjacent to intrusive rocks.

6.2.4 Mineralization

Three types of mineralization have been found on the property:

a) Vein-type:

Relatively flat lying quartz veins up to 1 meter thick carry sporadic pyrite-chalcopyrite-galena-hematite. Small local quartz veins and lenses within the pyroxenite unit carry pyrite-chalcopyrite.

b) Fracture controlled:

Malachite-chalcopyrite-magnetite occurs along fracture planes within the pyroxenite, diorite dykes, hybrid zone (in ultramafic xenoliths), and in quartz-feldspathic lenses of the feldspar porphyry.

c) Disseminated:

Chalcopyrite-pyrite, associated with strong carbonate-chlorite alteration, occurs as disseminations within Takla volcanics and in carbonate veinlets.

6.3 Rock Geochemistry (Figure 6)

For complete rock sample descriptions, see Appendix A.

Eleven rock samples returned significant assay results in copper (>1000 ppm) or gold (>300 ppb). For details, see Table 1 on the following page.

Sample 12206 was collected from an isolated outcrop in a talus filled creek valley north of the grid. Mineralization consisted of fracture-filled and disseminated chalcopyrite in a chlorite altered andesite.

A polished thin section was prepared from Sample 12206 (see Appendix D). The section was described as a crushed, altered porphyritic andesite/latite. Plagioclase and chloritized mafic phenocrysts are set in a "crushed" groundmass of sericitic plagioclase, chlorite altered mafics, and introduced K-feldspar. Opaques (5%) include fracture controlled and disseminated pyrite/chalcopyrite, magnetite as diffuse clusters, and specular hematite associated with chalcopyrite.

<u>Sample</u>	<u>Type</u>	<u>Width</u> <u>(m)</u>	<u>Cu</u> <u>(ppm)</u>	<u>Au</u> <u>(ppb)</u>	<u>Description</u>
12357	Select	-	2.3%	14000	Diorite subcrop with pyrite blebs and 2% chalcopyrite. 47.5 ppm Ag.
12358	Chip	0.3	2231	7	Hybrid unit with malachite stain.
12360	Select	-	2.7%	4070	Hybrid unit hosting flat-lying quartz veins mineralized with chalcopyrite and tetrahedrite. 79 ppm Ag.
12367	Chip	3.0	3261	< 5	Shear zone in malachite stain pyroxenite.
12477	Select	-	665	366	Pyroxenite with quartz veinlets.
12023	Chip	1.0	1440	202	Rusty andesite with trace chalcopyrite.
12025	Chip	2.0	11961	1100	Andesite porphyry with pyrite blebs and malachite/ azurite stain.
12201	Chip	0.5	272	398	Lapilli tuff with moderate ankerite alteration and disseminated pyrite/ pyrrhotite up to 10%
12203	Panel	2m ²	2.4%	99	Pyroxenite with strong malachite and limonite stain.
12205	Chip	2.0	1925	< 5	Quartz-feldspar dyke with moderate malachite/ azurite stain.
12206	Chip	1.0	1744	82	
12206	(reassayed)		2202	117	

6.4 Soil Geochemistry (Figures 7 and 8)

Summary Statistics:

	Copper	Gold
Range	8 to 1207 ppm	2.5 to 560 ppb
Mean	224.47	24.78
Standard deviation	268.77	78.30
Background	8 to 149 ppm	2.0 to 5 ppb
Low Anomalous	150 to 299 ppm	6 to 17 ppb
Medium Anomalous	300 to 599 ppm	18 to 29 ppb
High Anomalous	600+ ppm	30+ ppb

The correlation coefficient chart (Appendix C) shows no significant correlations between gold and other elements. Copper shows a strong correlation with molybdenum, a moderate correlation with silver, and a weak correlation with nickel.

The northeast area of the grid is overlain by a consistent copper anomaly measuring approximately 1500 by 600 meters. The anomaly is open to the north, east, and west, and is cut off sharply to the southwest at the contact between pyroxenite and diorite/hybrid units. Outcrop in the area consists of diorite, hybrid rocks, and andesites. The few small zones with malachite/azurite staining which were found in outcrop are not considered to be the mineralization which is the source of the size, consistency, and strength of the existing anomaly.

Eleven significant gold values in soils are clustered in the northeast area of the grid and are coincident with the copper anomaly described above. Highly anomalous values include 66, 70, 166, 204, 250, 370, 414, 436, and 560 ppb. Outcrop in the area is ankerite altered andesite with local zones of pyrite, chalcopyrite, malachite mineralization. Gold values from two rocks sampled in the area were 1100 and 1440 ppb. Six anomalous gold values up to 260 ppb occur along the west part of line 100+00N.

7. DISCUSSION

Although approximately 60% of the claim area has yet to be investigated, the 1992 exploration program has outlined targets which indicate good potential for extensive porphyry mineralization.

The first target is at rock sample site 12206. Fracture-filled and disseminated chalcopyrite mineralization is in a chlorite-sericite/K-feldspar altered porphyritic andesite. The outcrop is in a talus filled valley, and the general area has yet to be investigated.

The second area is a large, open ended copper/gold soil anomaly. Copper shows moderate to strong correlation with molybdenum and silver, both of which are often associated with porphyry deposits.

8. CONCLUSIONS

The ABE property has potential to host a porphyry style copper/gold deposit because:

- * it lies within the Mesozoic Quesnel Belt which hosts several significant porphyry copper/gold deposits;
- * the geological environment, which includes diorite, monzonite, and pyroxenite stocks intruding Takla Group volcanic rocks, is favorable;
- * porphyry style copper mineralization has been located in outcrop; and
- * a large copper/gold soil anomaly, yet to be fully outlined, indicates the presence of buried mineralization.

9. RECOMMENDATIONS

- a) Perform a total of approximately 25 line kilometers of induced polarization and magnetic geophysics over both target areas.

Contingent on favorable results from these IP surveys, further work would consist of diamond drilling to test the targets at depth.

- b) Perform reconnaissance geological mapping and rock sampling over the unmapped portions of the property.

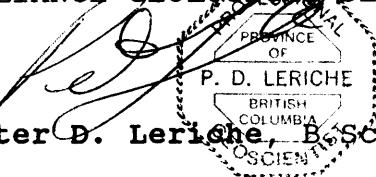
Contingent on favorable results from the mapping and sampling, further work would consist of detailed mapping, soil sampling, and IP surveys to establish drill targets.

CERTIFICATE

I, PETER D. LERICHE, of 3125 West 12th Avenue, Vancouver, B.C., V6K 2R6, do hereby state that:

1. I am a graduate of McMaster University, Hamilton, Ontario, with a Bachelor of Science Degree in Geology, 1980.
2. I am registered as a member in good standing with the Association of Professional Engineers and Geoscientists of British Columbia.
3. I am a Fellow in good standing with the Geological Association of Canada.
4. I have actively pursued my career as a geologist for twelve years in British Columbia, Ontario, the Yukon and Northwest Territories, Montana, Oregon, Alaska, Arizona, Nevada and California.
5. The information, opinions, and recommendations in this report are based on fieldwork carried out under my direction, and on published and unpublished literature. I visited the subject property during July 1992.
6. I have no interest, direct or indirect, in the subject claims or the securities of Swannell Minerals Corporation or Major General Resources Ltd, nor do I expect to receive any.
7. I consent to the use of this report only in its entirety in a Prospectus or Statement of Material Facts for the purpose of private or public financing.

RELIANCE GEOLOGICAL SERVICES INC.


P. D. LERICHE
PROVINCE OF
BRITISH COLUMBIA
GEOSCIENCE
Peter D. Leriche, B.Sc., P.Geo.

Dated at North Vancouver, B.C., this 23rd day of February 1993.

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55 pp.

ITEMIZED COST STATEMENT
ABE PROJECT (J770)

Project preparation	\$ 550
Mobilization and demobilization (includes transportation, and wages)	\$ 5,080
Consulting	\$ 500
Field Crew:	
Supervision	\$ 650
Project Geologist \$ 325/day x 6 days	\$ 1,950
(A. Taylor: Jul 25,26,27, Aug 22,23, 26, 1992)	
Field Geologists \$ 275/day x 12 days	\$ 3,300
(D. Johannessen & R. Faulkner Jul 25,26,27, Aug 22,23, 26, 1992)	
Geotechnicians \$ 210/day x 6 days	\$ 1,260
(\$ 7,160 (B. Chore: Jul 25,26,27, 1992) (J. Fleishman: Aug 22,23, 26, 1992)	
Field Costs:	
Helicopter \$ 840/hr x 6.3 hrs	\$ 5,292
Communications \$ 50/day x 6 days	\$ 300
Expediting and freight	\$ 400
Food & accommodation \$ 75/day x 24 days	\$ 1,800
Supplies \$ 18/day x 24 days	\$ 432
Vehicle \$ 30/day x 6 days	\$ 180
	\$ 8,404
Assays & Analysis:	
157 soil samples @ \$14/sample	\$ 2,198
(Geochem/AA for Au + 30 element ICP)	
35 rock samples @ \$17/sample	\$ 595
(FA/AA for Au and 30 element ICP)	
Thin section analysis	\$ 150
41 soil prep @ 1.50/sample	\$ 61
	\$ 3,004
Report:	
Writing, editing, map prep, processing, binding, copying	\$ 1,875
Administration, incl overhead and profit	\$ 2,657
Sub-total	\$ 29,230
plus 7% G.S.T.	\$ 2,046
TOTAL	\$ 31,276

APPENDIX A

Rock Sample Descriptions

ROCK SAMPLE DESCRIPTIONS

ABE Claims

SAMPLE NUMBER	TYPE	WIDTH (meters)	DESCRIPTION
12023	chip	1.0	Below old trench. Rusty ankeritic andesite, trace malachite and chalcopyrite.
12024	chip	2.0	In old trench. Rusty ankerite. Trace malachite with calcite veinlets.
12025	chip	2.0	Malachite-azurite stained andesite porphyry. Blobs of pyrite up to 5% occur sporadically.
12026	chip	1.0	Augite porphyry with minor malachite. Trace chalcopyrite, magnetite, hematite, and calcite veinlets.
12201	chip	.5	10590N, 12550E. Lapilli tuff with moderate ankeritic alteration adjacent to shear zone. Weathers black with 10% local disseminated pyrite/pyrrhotite.
12202	chip	.5	10450N, 11650E. Rusty weathering, pyritic pyroxene porphyry.
12203	panel	2 m ²	10200N, 11010E. Rusty subcrop. Limonite-malachite debris in pyroxenite adjacent to feldspar porphyry dyke.
12205	chip	2.0	10800N, 11150E, cliff. Quartz-feldspar dyke intruding feldspar porphyry. Moderate malachite-azurite stain.
12206	chip	1.0	11000N, 12000E, in poorly exposed creekcut. Carbonate-chlorite rich andesite with chalcopyrite-carbonate vein and disseminated pyrite.

ROCK SAMPLE DESCRIPTIONS

ABE Claims

SAMPLE NUMBER	TYPE	WIDTH (meters)	DESCRIPTION
12357	select		9995N, 9960E, 1770 m, beside lake. Pyrite blebs with heavy malachite stain, 2% chalcopyrite, in diorite (subcrop)
12358	chip	.3	1075N, 9300E, 2000 m, ridgetop. Malachite stain in hybrid felsite/ultramafic unit adjacent to a porphyry dyke.
12359	chip	.15	10400N, 10205E, 1690 m. Three 1 cm quartz-ankerite veins with platy hematite visible, host diorite.
12360	select		10300N, 10700E, 1676 m. Hybrid unit with large ultramafic pyroxenite blocks. Limonitic area with reddish hematite stain. Outcrop is cut by flat-lying quartz veins containing visible chalcopyrite, tetrahedrite and hematite.
12361	chip	1.5	same as 12360
12362	select		10600N, 11100E, 1770 m. Diorite with 1 mm veinlets of pyrite next to feldspar porphyry.
12363	select		10600N, 11170E, 1825 m. Semi-massive fine-grained magnetite vein next to feldspar porphyry.
12364	chip	1.0	10425N, 11475E, 1955 m. Gossanous pyroxenite with malachite stain, hybrid unit.
12365	chip	1.0	9650N, 10850E, 1915 m, cliff. Two 1 cm quartz veinlets with galena and pyrite in feldspar porphyry.

ROCK SAMPLE DESCRIPTIONS

ABE Claims

SAMPLE NUMBER	TYPE	WIDTH (meters)	DESCRIPTION
12366	chip	1.0	9500N, 10900E, 2000 m. Two 20 cm flat-lying quartz veins in silicic and ankeritic diorite. Trace galena and pyrite.
12367	chip	3.0	10 m west of peak, 2040 m. Rusty pyroxenite with visible malachite in a shear zone.
12368	chip	1.0	West of 12367. Limonitic shear zone, ankerite with pyroxenite and hematite.
12369	chip	3.0	9600N, 11045E, 2032 m, ridge. Chip across dioritic unit cut by three horizontal quartz veins. No obvious mineralization.
12370	chip	2.0	9340N, 11400E, 1850 m. Ankeritic diorite with multiple quartz veins mineralized with galena.
12371	chip	2.0	9350N, 11500E, 1830 m. Medium-grained, dark green pyroxenite.
12372	chip	3.0	Limonitic/ankeritic horizon in Takla tuffs with disseminated 5% pyrite.
12427	select		Pyroxenite talus with disseminated pyrite. Small outcrops are scattered about the area.
12428	select		Reddish to bright-orange weathering intrusive with specks of magnetite.
12429	select		Pyritic siliceous fine-grained volcanic. Talus.

ROCK SAMPLE DESCRIPTIONS

ABE Claims

SAMPLE NUMBER	TYPE	WIDTH (meters)	DESCRIPTION
12476	select		1850 m, ridge. Andesite feldspar porphyry dyke. Slightly silicified. 2 to 3% pyrite as blebs. Minor sphalerite(?) intergrown with some pyrite blebs.
12477	select		Same ridge as 12476. Pyroxenite with limonite stains and 30% quartz veinlets. 5 to 10% magnetite as large blebs.
12478	chip	2.0	On ridge near eastern boundary of Abe 5. Three meter wide zone of anastomosing quartz ankerite veinlets to 10 mm in augite porphyry. Minor pyrite as blebs to 1 mm.
12479	select		On spur of ridge near eastern boundary of Abe 5. Augite porphyry (Takla). Clay altered with pods of silicification. 3 to 20% boxwork, some relict pyrite.
12480	select		On north-south ridge in north-eastern part of claim group at 1900 m. Fine-grained black basalt. Limonite stained and clay altered. Some small silicified patches. 5 to 10% boxwork, 2 to 3% pyrite.
12481	select		On small knob directly east of lake at 1700 m. Ankeritized zone in fine-grained andesite. Slightly silicified.

APPENDIX B

Analytical Reports and Techniques

R E P O R T S U M M A R Y

Report:[9200579 R]

A N A L Y T I C A L R E P O R T

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Origin

Inception Date:[Jul 31, 1992]

Client:[269	Reliance Geological Services Ltd.]
Contact:[Peter Leriche]
Project:[0	AB]
Amount/Type:[25	Rock]
[]

Analytical Requisition

Geochemical:[ICP(AgR)30] ICP:[30]
Assay:[Au(FA/AAS 20g)]
Comments:[None]

Delivery Information

Reporting Date:[Aug 11, 1992]

Principal Destination (Hardcopy,Fascimile,Invoice)

Company:[Reliance Geological Services Ltd.]]
Address:[241 East 1st Street]]
City/Province:[North Vancouver, BC]]
Country/Postal:[V7L 1B4]]
Attention:[Peter Leriche]]
Fascimile:[(604)988-4653]]

Secondary Destination (Hardcopy)

Company:[]]
Address:[]]
City/Province:[]]
Country/Postal:[]]
Attention:[]]
Fascimile:[]]

1 data pages in this report.

Approved by:



B.C. Certified Assayers

iPL CODE: 920811-15:31:11

Report: 9200579 R	Reliance Geological Services Ltd.	Project: AB								Page 1 of 1	Section 1 of 2						
Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
12357	Rock	14m	47.5	2.3%	9	64	<5	<5	<3	8	<10	<2	1.3	32	33	13	10
12358	Rock	7	1.2	2231	<2	16	<5	<5	<3	1	<10	<2	<0.1	16	25	6	<5
12359	Rock	10	<0.1	57	<2	31	<5	<5	<3	1	<10	<2	<0.1	13	6	249	<5
12360	Rock	4070	79.3	2.7%	539	81	<5	7	<3	28	<10	87	2.8	27	16	17	<5
12361	Rock	804	4.8	1018	40	35	<5	<5	<3	26	<10	<2	0.2	14	8	59	<5
12362	Rock	<5	<0.1	132	<2	33	<5	<5	<3	2	<10	<2	<0.1	23	42	163	<5
12363	Rock	33	<0.1	140	2	19	34	<5	<3	49	<10	17	0.2	43	184	33	65
12364	Rock	<5	<0.1	182	<2	31	<5	<5	<3	3	<10	<2	<0.1	25	133	28	<5
12365	Rock	5	33.0	38	6312	47	<5	<5	<3	99	<10	65	2.1	8	7	39	<5
12366	Rock	44	14.6	336	988	21	<5	<5	<3	4	<10	24	0.7	6	6	327	<5
12367	Rock	<5	0.4	3261	10	30	<5	9	<3	2	<10	<2	<0.1	35	71	27	<5
12368	Rock	8	0.8	853	13	24	13	<5	5	12	<10	<2	0.5	98	286	111	<5
12369	Rock	21	1.0	40	135	90	<5	6	<3	2	<10	<2	2.8	7	19	633	<5
12370	Rock	<5	0.3	30	207	410	<5	<5	<3	2	<10	<2	8.1	9	8	204	<5
12371	Rock	<5	<0.1	7	2	23	<5	23	3	2	<10	<2	<0.1	41	165	5	<5
12372	Rock	<5	0.1	26	8	46	6	6	<3	7	<10	<2	<0.1	11	34	15	<5
12427	Rock	15	0.5	701	3	10	<5	6	<3	1	<10	<2	<0.1	57	81	33	<5
12428	Rock	<5	<0.1	5	3	22	<5	<5	<3	2	<10	<2	<0.1	5	3	137	<5
12429	Rock	<5	0.2	14	2	24	<5	<5	<3	3	<10	<2	<0.1	13	14	30	<5
12476	Rock	<5	0.1	195	<2	4	<5	<5	<3	18	<10	<2	<0.1	12	6	64	6
12477	Rock	366	2.5	665	<2	32	40	<5	<3	13	<10	5	<0.1	21	54	<2	<5
12478	Rock	12	<0.1	57	4	108	<5	5	<3	3	<10	<2	0.4	25	77	432	<5
12479	Rock	8	<0.1	29	2	30	<5	<5	<3	3	<10	<2	<0.1	16	26	10	<5
12480	Rock	<5	0.2	130	<2	10	<5	<5	<3	<1	<10	<2	<0.1	14	18	65	<5
12481	Rock	5	<0.1	29	<2	43	<5	<5	4	16	<10	<2	0.1	25	28	226	<5

Minimum Detection
Maximum Detection
Method

5	0.1	1	2	1	5	5	10000	100.0	20000	20000	10000	1000	10000	10	2	0.1	1	1	2	5	
FA/AAS	ICP	ICP	ICP	ICP	ICP	ICP															

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

Report: 9200579 R Reliance Geological Services Ltd. Project: AB Page 1 of 1 Section 2 of 2

Sample Name	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti %	A1 %	Ca %	Fe %	Mg %	K %	Na %	P %
12357	170	138	494	<2	14	1	6	0.08	0.99	0.47	9.4%	1.34	0.03	0.04	0.08
12358	187	11	254	<2	12	<1	3	0.03	0.69	1.87	1.27	1.30	0.01	0.03	0.01
12359	70	38	966	4	65	<1	4	0.01	0.76	2.71	3.26	0.75	0.25	0.05	0.11
12360	133	22	2057	<2	156	4	2	<0.01	0.81	2.22	7.0%	1.00	0.09	0.02	0.06
12361	91	33	1887	4	116	6	3	0.01	0.97	2.27	4.57	0.90	0.16	0.05	0.06
12362	48	140	420	<2	70	1	4	0.20	2.56	1.18	3.68	2.56	0.93	0.04	0.04
12363	17	110	242	<2	6	10	<1	0.03	0.16	0.07	33%	0.16	0.41	0.02	0.01
12364	102	185	481	2	66	1	3	0.12	1.63	1.30	6.0%	1.67	0.12	0.06	0.34
12365	152	40	353	3	57	9	1	0.08	0.44	0.90	2.19	0.48	0.14	0.08	0.05
12366	178	23	484	2	17	8	1	<0.01	0.27	0.57	1.36	0.12	0.06	0.06	0.04
12367	386	70	257	<2	11	2	7	0.07	0.55	1.60	3.46	1.30	0.31	0.09	0.01
12368	108	268	531	<2	30	3	7	0.08	0.83	2.38	11%	1.42	0.82	0.11	0.08
12369	152	20	5171	4	82	6	6	0.01	0.48	4.47	2.77	1.61	0.26	0.03	0.03
12370	95	65	1215	4	64	7	4	0.01	0.34	2.32	3.23	0.36	0.11	0.08	0.07
12371	897	51	539	<2	10	1	6	0.06	0.66	2.26	4.93	4.65	0.01	0.02	<0.01
12372	123	41	273	2	47	3	2	0.16	1.39	1.22	4.28	0.30	0.10	0.04	0.05
12427	237	44	144	<2	40	2	2	0.11	0.66	0.63	3.53	0.91	0.11	0.04	0.02
12428	64	16	287	9	22	3	1	<0.01	0.83	0.87	1.37	0.41	0.27	0.06	0.07
12429	70	65	195	2	105	2	2	0.18	1.12	0.97	3.51	0.44	0.19	0.10	0.08
12476	40	48	46	7	54	7	1	0.14	0.28	0.25	3.73	0.11	0.17	0.10	0.12
12477	5	1130	293	<2	2	10	2	0.09	0.24	0.04	33%	0.07	0.03	0.02	0.01
12478	59	90	1296	<2	55	1	13	<0.01	1.35	7.40	5.1%	2.55	0.20	0.04	0.06
12479	96	193	142	<2	7	5	11	0.24	1.56	0.11	7.2%	2.21	0.09	0.06	0.07
12480	40	158	148	<2	54	3	8	0.29	0.83	1.00	4.82	0.72	0.22	0.07	0.03
12481	39	22	1259	<2	144	1	13	<0.01	0.52	8.00	4.40	2.15	0.24	0.02	0.07

Minimum Detection	1	2	1	2	1	1	1	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01
Maximum Detection	10000	10000	10000	10000	10000	10000	10000	1.00	5.00	10.00	5.00	10.00	10.00	5.00	5.00
Method	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP						

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

R E P O R T S U M M A R Y

Report:[9200740 R]

A N A L Y T I C A L R E P O R T

=====

Origin

Inception Date:[Sep 09, 1992]

Client:[269	Reliance Geological Services Ltd.]
Contact:[Peter Leriche]
Project:[0	770 ABC]
Amount/Type:[10	Rock]
[

Analytical Requisition

Geochemical:[ICP(AqR)30] ICP:[30]
Assay:[Au(FA/AAS 20g)	
Comments:[None]

Delivery Information

Reporting Date:[Sep 11, 1992]

Principal Destination (Hardcopy, Fascimile, Invoice)

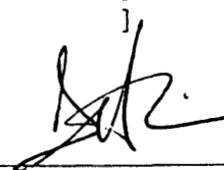
Company:[Reliance Geological Services Ltd.]]
Address:[241 East 1st Street]]
City/Province:[North Vancouver, BC]]
Country/Postal:[V7L 1B4]]
Attention:[Peter Leriche]]
Fascimile:[604/988-4653]]

Secondary Destination (Hardcopy)

Company:[]]
Address:[]]
City/Province:[]]
Country/Postal:[]]
Attention:[]]
Fascimile:[]]

1 data pages in this report.

Approved by:



B.C. Certified Assayers

iPL CODE: 920911-12:26:07

Report: 9200740 R	Reliance Geological Services Ltd.	Project: 770 ABC										Page 1 of 1	Section 1 of 2				
		Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm
12023	Rock	1440	0.5	202	<2	543	14	<5	<3	3	<10	<2	1.4	25	15	85	<5
12024	Rock	10	0.2	498	3	105	<5	6	<3	4	<10	<2	0.6	20	59	347	<5
12025	Rock	1100	3.8	11961	7	53	<5	<5	<3	4	<10	<2	2.9	116	149	56	<5
12026	Rock	193	0.2	585	<2	77	<5	<5	<3	5	<10	<2	0.6	55	112	46	<5
12106	Rock	717	7.3	10	235	1	<5	<5	<3	5	<10	20	<0.1	5	6	52	<5
12201	Rock	398	75.5	272	156	107	259	<5	<3	9	<10	11	<0.1	39	51	<2	<5
12202	Rock	6	0.7	529	<2	66	5	<5	<3	4	<10	<2	<0.1	21	63	31	<5
12203	Rock	99	19.7	2.4%	14	166	7	<5	<3	227	<10	<2	<0.1	86	346	7	6
12205	Rock	<5	<0.1	1925	<2	12	<5	<5	<3	5	<10	<2	0.1	57	38	11	<5
12206	Rock	82	0.8	1744	<2	60	<5	<5	<3	4	<10	<2	0.2	32	106	<2	<5

Minimum Detection	5	0.1	1	2	1	5	5	3	1	10	2	0.1	1	1	2	5
Maximum Detection	10000	100.0	20000	20000	20000	10000	1000	10000	1000	10000	10000.0	10000	10000	10000	10000	1000
Method	FA/AAS	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

Sample Name	Project: 770 ABC												Page	1 of 1	Section	2 of 2
	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti %	A1 %	Ca %	Fe %	Mg %				
12023	40	104	205	3	17	2	6	0.10	0.82	0.20	4.40	0.71	0.22	0.04	0.09	
12024	148	81	1929	<2	101	2	14	<0.01	1.28	7.84	5.8%	4.16	0.28	0.01	0.07	
12025	238	216	1055	<2	47	3	13	0.14	3.31	2.40	8.1%	5.34	0.13	0.02	0.11	
12026	306	179	2750	<2	64	2	20	0.03	3.37	5.57	10%	4.43	0.22	0.01	0.08	
12106	206	12	33	3	8	3	<1	<0.01	0.21	0.06	2.05	0.03	0.26	0.02	0.01	
12201	157	40	123	<2	2	2	2	<0.01	0.28	0.02	16%	0.21	0.01	0.01	<0.01	
12202	154	105	913	<2	79	2	4	0.15	1.79	0.83	4.88	2.05	0.13	0.05	0.11	
12203	28	113	534	5	36	2	1	0.01	0.49	2.73	14%	0.56	0.07	0.01	1.29	
12205	53	16	136	<2	30	<1	1	0.02	0.76	0.35	0.57	0.46	0.12	0.10	0.06	
12206	356	181	1541	<2	14	1	18	0.08	4.77	1.30	8.8%	6.42	0.05	0.01	0.05	

Minimum Detection 1 2 1 2 1 1 1 1 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01
 Maximum Detection 10000 10000 10000 10000 10000 10000 10000 10000 1.00 5.00 10.00 5.00 10.00 10.00 5.00 5.00
 Method ICP
 ---No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

R E P O R T S U M M A R Y

Report:[9200779 R]

A N A L Y T I C A L R E P O R T

=====

Origin

Inception Date:[Sep 16, 1992]

Client:[269	Reliance Geological Services Ltd.]
Contact:[Peter Leriche]
Project:[0	770 Abe]
Amount/Type:[1	Rock]
	[]

Analytical Requisition

Geochemical:[ICP(AqR)30] ICP:[30]
Assay:[Au(FA/AAS 30g)	
Comments:[Re:9200740]

Delivery Information

Reporting Date:[Sep 18, 1992]

Principal Destination (Hardcopy,Fascimile,Invoice)

Company:[Reliance Geological Services Ltd.]]
Address:[241 East 1st Street]]
City/Province:[North Vancouver, BC]]
Country/Postal:[V7L 1B4]]
Attention:[Peter Leriche]]
Fascimile:[(604)988-4653]]

Secondary Destination (Hardcopy)

Company:[]]
Address:[]]
City/Province:[]]
Country/Postal:[]]
Attention:[]]
Fascimile:[]]

1 data pages in this report.

Approved by:

B.C. Certified Assayers

iPL CODE: 920918-17:36:23

Report: 9200779 R Reliance Geological Services Ltd.

Project: 770 Abe

Page 1 of 1

Section 1 of 2

Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
12206	Rock	117	0.7	2202	<2	65	<5	<5	<3	3	<10	<2	<0.1	35	116	<2	<5

Minimum Detection
Maximum Detection
Method

2	0.1	1	2	1	5	5	3	1	10	2	0.1	1	1	2	5	
10000	100.0	20000	20000	20000	10000	1000	10000	1000	1000	10000	10000.0	10000	10000	10000	10000	1000
FA/AAS	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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Report: 9200779 R Reliance Geological Services Ltd.

Project: //0 Abe

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Sample Name	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti %	Al %	Ca %	Fe %	Mg %	K %	Na %	P %
12206	422	209	1725	<2	16	2	21	0.16	5.3%	1.52	9.8%	7.11	0.06	0.02	0.06

Minimum Detection 1 2 1 2 1 1 1 0.01 0.01 0.01 0.01 0.01 0.01 0.01
 Maximum Detection 10000 10000 10000 10000 10000 10000 10000 1.00 5.00 10.00 5.00 10.00 10.00 5.00 5.00
 Method ICP
 --=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

R E P O R T S U M M A R Y

Report:[9200566 R]

A N A L Y T I C A L R E P O R T
=====

Origin

Inception Date:[Jul 31, 1992]

Client:[269	Reliance Geological Services Ltd.]
Contact:[Peter Leriche]
Project:[0	AB]
Amount/Type:[158	Soil]
[]

Analytical Requisition

Geochemical:[ICP(AqR)30] ICP:[30]
Assay:[Au(FA/AAS 10g)	
Comments:[No Instructions]

Delivery Information

Reporting Date:[Aug 10, 1992]

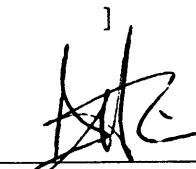
Principal Destination (Hardcopy, Fascimile, Invoice)

Company:[Reliance Geological Services Ltd.]]
Address:[241 East 1st Street]]
City/Province:[North Vancouver, BC]]
Country/Postal:[V7L 1B4]]
Attention:[Peter Leriche]]
Fascimile:[(604)988-4653]]

Secondary Destination (Hardcopy)

Company:[]]
Address:[]]
City/Province:[]]
Country/Postal:[]]
Attention:[]]
Fascimile:[]]

5 data pages in this report.

Approved by: 

B.C. Certified Assayers

iPL CODE: 920810-11:22:54

Report: 9200566 R Reliance Geological Services Ltd. Project: AB Page 1 of 5 Section 1 of 2

Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
AB92 94+00N 108+00E	Soil	<5	0.1	37	52	116	<5	13	4	2	<10	<2	0.1	32	124	151	<5
AB92 94+00N 109+00E	Soil	<5	<0.1	57	38	39	<5	10	<3	2	<10	<2	0.2	30	78	109	<5
AB92 94+00N 110+00E	Soil	<5	<0.1	87	52	45	<5	8	3	2	<10	<2	<0.1	30	88	104	<5
AB92 94+00N 111+00E	Soil	<5	0.4	107	259	59	<5	5	<3	4	<10	4	0.3	27	35	996	<5
AB92 94+00N 112+00E	Soil	<5	<0.1	39	311	32	<5	7	<3	2	<10	<2	<0.1	24	63	111	<5
AB92 94+00N 113+00E	Soil	<5	<0.1	41	99	46	<5	7	<3	3	<10	<2	<0.1	28	75	145	<5
AB92 94+00N 114+00E	Soil	<5	0.2	69	115	68	<5	9	4	2	<10	<2	<0.1	35	92	200	<5
AB92 94+00N 115+00E	Soil	<5	<0.1	18	25	52	<5	16	<3	2	<10	<2	<0.1	57	156	36	<5
AB92 94+00N 116+00E	Soil	<5	<0.1	24	3	94	<5	8	<3	4	<10	<2	<0.1	30	69	280	<5
AB92 94+00N 117+00E	Soil	<5	<0.1	37	35	71	<5	15	<3	4	<10	<2	<0.1	46	125	35	<5
AB92 94+00N 118+00E	Soil	<5	<0.1	23	48	42	<5	11	5	2	<10	<2	<0.1	41	106	26	<5
AB92 94+00N 119+00E	Soil	<5	<0.1	15	22	63	<5	11	<3	3	<10	<2	<0.1	35	77	16	<5
AB92 94+00N 120+00E	Soil	<5	<0.1	42	25	77	<5	9	<3	7	<10	<2	<0.1	37	85	38	<5
AB92 94+00N 121+00E	Soil	<5	0.1	36	31	70	<5	15	3	3	<10	<2	<0.1	46	96	30	<5
AB92 94+00N 122+00E	Soil	<5	<0.1	30	12	54	<5	9	<3	11	<10	<2	<0.1	32	67	36	<5
AB92 94+00N 123+00E	Soil	<5	0.2	133	19	68	<5	8	<3	7	<10	<2	<0.1	34	100	86	<5
AB92 96+00N 108+00E	Soil	6	<0.1	43	51	47	<5	9	<3	4	<10	<2	0.1	37	106	88	<5
AB92 96+00N 110+00E	Soil	<5	<0.1	28	17	54	<5	17	<3	2	<10	<2	<0.1	42	152	68	<5
AB92 96+00N 111+00E	Soil	<5	<0.1	31	110	56	<5	<5	<3	2	<10	<2	<0.1	25	59	106	<5
AB92 96+00N 112+00E	Soil	<5	<0.1	27	146	41	<5	8	<3	2	<10	<2	0.1	25	68	91	<5
AB92 96+00N 113+00E	Soil	<5	<0.1	29	56	51	<5	10	<3	2	<10	<2	<0.1	34	105	82	<5
AB92 96+00N 114+00E	Soil	<5	<0.1	40	86	68	<5	11	3	6	<10	<2	<0.1	35	108	114	<5
AB92 96+00N 115+00E	Soil	<5	<0.1	21	<2	42	<5	7	<3	2	<10	<2	<0.1	25	85	11	<5
AB92 96+00N 116+00E	Soil	<5	<0.1	10	21	52	<5	6	<3	3	<10	<2	<0.1	17	47	25	<5
AB92 96+00N 117+00E	Soil	<5	<0.1	8	31	47	<5	7	<3	3	<10	<2	<0.1	17	50	21	<5
AB92 96+00N 118+00E	Soil	<5	<0.1	29	12	40	<5	12	<3	3	<10	<2	<0.1	34	100	14	<5
AB92 96+00N 119+00E	Soil	6	<0.1	22	12	49	<5	9	<3	5	<10	<2	<0.1	26	70	21	<5
AB92 96+00N 120+00E	Soil	<5	0.1	28	17	41	<5	9	4	4	<10	<2	<0.1	28	79	7	<5
AB92 96+00N 121+00E	Soil	<5	<0.1	33	13	46	<5	10	<3	3	<10	<2	<0.1	25	60	18	<5
AB92 96+00N 122+00E	Soil	<5	<0.1	41	17	41	<5	11	<3	3	<10	<2	<0.1	26	73	15	<5
AB92 96+00N 123+00E	Soil	<5	0.2	45	12	32	<5	10	<3	2	<10	<2	<0.1	27	61	17	<5
AB92 98+00N 108+00E	Soil	<5	0.5	95	136	57	<5	11	3	4	<10	<2	<0.1	32	80	161	<5
AB92 98+00N 109+00E	Soil	<5	0.2	102	45	43	<5	12	<3	3	<10	<2	<0.1	42	106	89	<5
AB92 98+00N 110+00E	Soil	6	0.4	101	110	86	<5	13	<3	3	<10	<2	0.1	45	124	47	<5
AB92 98+00N 111+00E	Soil	<5	<0.1	34	20	42	<5	11	<3	3	<10	<2	<0.1	32	98	81	<5
AB92 98+00N 112+00E	Soil	<5	0.1	30	22	51	<5	<5	3	3	<10	<2	<0.1	37	90	103	<5
AB92 98+00N 113+00E	Soil	38	<0.1	32	36	73	<5	<5	<3	5	<10	<2	<0.1	28	50	85	<5
AB92 98+00N 114+00E	Soil	<5	<0.1	71	37	75	<5	10	3	8	<10	<2	<0.1	33	89	86	<5
AB92 98+00N 115+00E	Soil	<5	<0.1	37	40	43	<5	7	<3	8	<10	<2	<0.1	27	76	64	<5
Minimum Detection		5	0.1	1	2	1	5	5	3	1	10	2	0.1	1	1	2	5
Maximum Detection		10000	100.0	20000	20000	20000	10000	1000	10000	1000	10000	10000	0.1	10000	10000	10000	10000
Method		FA/AAS	ICP	ICP													

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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Report: 9200566 R Reliance Geological Services Ltd.

Project: AB

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Section 2 of

Report: 9200566 R Reliance Geological Services Ltd.

Project: AB

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Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
AB92 98+00N 116+00E	Soil	<5	<0.1	38	45	34	<5	8	<3	3	<10	<2	<0.1	31	83	49	<5
AB92 98+00N 117+00E	Soil	18	<0.1	16	3	69	<5	8	<3	3	<10	<2	<0.1	31	86	37	<5
AB92 98+00N 118+00E	Soil	<5	0.2	81	9	76	<5	7	<3	8	<10	<2	<0.1	49	77	44	<5
AB92 98+00N 119+00E	Soil	<5	<0.1	643	5	45	<5	<5	<3	16	<10	<2	<0.1	43	63	53	<5
AB92 98+00N 120+00E	Soil	12	0.1	701	3	50	<5	5	<3	23	<10	<2	<0.1	37	84	77	<5
AB92 98+00N 121+00E	Soil	10	0.1	987	10	66	<5	<5	<3	21	<10	<2	<0.1	72	87	65	<5
AB92 98+00N 122+00E	Soil	<5	0.1	364	3	75	<5	9	<3	22	<10	<2	<0.1	52	90	98	<5
AB92 98+00N 123+00E	Soil	<5	0.2	388	<2	99	<5	<5	<3	11	<10	<2	<0.1	41	65	100	<5
AB92 100+00N 75+00E	Soil	14	0.1	55	6	50	<5	<5	4	7	<10	<2	<0.1	19	34	202	<5
AB92 100+00N 76+00E	Soil	6	<0.1	87	4	67	<5	5	<3	3	<10	<2	<0.1	24	53	236	<5
AB92 100+00N 77+00E	Soil	22	0.3	91	11	78	<5	<5	<3	5	<10	<2	<0.1	29	57	199	<5
AB92 100+00N 78+00E	Soil	<5	0.1	135	18	106	13	<5	<3	13	<10	<2	<0.1	45	45	165	153
AB92 100+00N 79+00E	Soil	94	0.5	210	3	76	<5	6	4	7	<10	<2	<0.1	47	118	260	<5
AB92 100+00N 80+00E	Soil	64	0.4	338	<2	75	<5	6	<3	12	<10	<2	<0.1	67	131	464	<5
AB92 100+00N 81+00E	Soil	260	1.3	194	64	75	<5	5	<3	42	<10	<2	<0.1	48	93	548	<5
AB92 100+00N 82+00E	Soil	28	<0.1	130	<2	60	<5	<5	<3	6	<10	<2	<0.1	31	42	676	<5
AB92 100+00N 83+00E	Soil	12	<0.1	83	2	66	<5	<5	<3	4	<10	<2	<0.1	24	33	374	<5
AB92 100+00N 84+00E	Soil	22	<0.1	166	<2	66	<5	<5	<3	7	<10	<2	<0.1	25	22	495	<5
AB92 100+00N 85+00E	Soil	<5	<0.1	40	10	28	<5	9	<3	2	<10	<2	<0.1	24	72	134	<5
AB92 100+00N 86+00E	Soil	<5	<0.1	138	<2	23	<5	5	<3	1	<10	<2	0.1	22	48	150	<5
AB92 100+00N 87+00E	Soil	<5	<0.1	276	2	23	<5	5	<3	1	<10	<2	<0.1	24	38	203	<5
AB92 100+00N 88+00E	Soil	<5	<0.1	103	<2	24	<5	5	<3	1	<10	<2	<0.1	17	32	62	<5
AB92 100+00N 89+00E	Soil	8	<0.1	83	<2	32	5	5	<3	1	<10	<2	<0.1	16	27	80	<5
AB92 100+00N 90+00E	Soil	<5	<0.1	172	<2	17	<5	6	<3	1	<10	<2	<0.1	20	39	48	<5
AB92 100+00N 91+00E	Soil	<5	<0.1	413	<2	21	<5	7	<3	1	<10	<2	<0.1	44	61	76	<5
AB92 100+00N 92+00E	Soil	<5	0.2	117	<2	37	<5	<5	4	2	<10	<2	<0.1	33	47	491	<5
AB92 100+00N 93+00E	Soil	14	<0.1	95	<2	37	<5	<5	<3	2	<10	<2	<0.1	20	34	110	<5
AB92 100+00N 94+00E	Soil	<5	<0.1	56	<2	26	<5	<5	<3	1	<10	<2	<0.1	14	22	107	<5
AB92 100+00N 95+00E	Soil	<5	<0.1	118	2	47	<5	5	<3	2	<10	<2	<0.1	21	37	117	<5
AB92 100+00N 96+00E	Soil	98	<0.1	138	<2	33	<5	<5	<3	2	<10	<2	<0.1	19	27	129	<5
AB92 100+00N 97+00E	Soil	<5	<0.1	50	<2	34	<5	<5	<3	2	<10	<2	<0.1	17	31	84	<5
AB92 100+00N 98+00E	Soil	<5	<0.1	31	2	18	<5	<5	<3	3	<10	<2	<0.1	10	16	133	<5
AB92 100+00N 99+00E	Soil	<5	<0.1	37	<2	22	5	<5	<3	3	<10	<2	<0.1	9	9	132	<5
AB92 100+00N 100+00E	Soil	<5	<0.1	55	3	15	<5	<5	<3	2	<10	<2	<0.1	9	8	85	<5
AB92 100+00N 101+00E	Soil	<5	<0.1	124	<2	48	<5	<5	<3	3	<10	<2	<0.1	23	28	148	<5
AB92 100+00N 102+00E	Soil	36	<0.1	77	<2	44	<5	<5	<3	3	<10	<2	<0.1	18	34	108	<5
AB92 100+00N 103+00E	Soil	14	0.5	201	61	35	<5	10	<3	9	<10	<2	<0.1	46	68	193	<5
AB92 100+00N 104+00E	Soil	<5	0.1	146	28	109	<5	6	3	3	<10	<2	0.4	27	50	97	<5
AB92 100+00N 105+00E	Soil	<5	0.7	173	163	85	<5	7	<3	16	<10	<2	0.2	31	60	104	<5

Minimum Detection

Maximum Detection

Method

5	0.1	1	2	1	5	5	3	1	10	2	0.1	1	1	2	5	
10000	100.0	20000	20000	20000	10000	1000	10000	1000	10000	10000	ICP	ICP	ICP	ICP	10000	
FA/AAS	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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Report: 9200566 R Reliance Geological Services Ltd.

Project: AB

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Report: 9200566 R Reliance Geological Services Ltd. Project: AB Page 3 of 5 Section 1 of 2

Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
AB92 100+00N 106+00E	Soil	<5	0.9	1188	85	131	<5	<5	<3	7	<10	<2	<0.1	77	54	183	<5
AB92 100+00N 107+00E	Soil	<5	0.5	76	181	89	<5	19	<3	7	<10	<2	<0.1	43	110	180	<5
AB92 100+00N 108+00E A	Soil	<5	0.1	55	56	55	<5	15	<3	3	<10	<2	<0.1	38	107	184	<5
AB92 100+00N 108+00E B	Soil	<5	0.2	56	69	67	<5	15	<3	3	<10	<2	<0.1	39	114	182	<5
AB92 100+00N 109+00E	Soil	<5	<0.1	97	43	51	<5	9	4	3	<10	<2	<0.1	38	112	76	<5
AB92 100+00N 110+00E	Soil	<5	<0.1	26	25	31	<5	13	<3	2	<10	<2	<0.1	29	94	40	<5
AB92 100+00N 111+00E	Soil	<5	<0.1	23	12	67	<5	<5	<3	4	<10	<2	<0.1	47	63	182	<5
AB92 100+00N 112+00E	Soil	<5	<0.1	27	34	36	<5	8	<3	2	<10	<2	<0.1	35	92	53	<5
AB92 100+00N 113+00E	Soil	<5	<0.1	41	27	30	<5	9	<3	2	<10	<2	<0.1	30	76	44	<5
AB92 100+00N 114+00E	Soil	<5	<0.1	13	2	22	<5	9	<3	1	<10	<2	<0.1	28	68	6	<5
AB92 100+00N 115+00E	Soil	<5	<0.1	85	11	42	<5	13	<3	6	<10	<2	<0.1	46	102	23	<5
AB92 100+00N 116+00E	Soil	<5	<0.1	334	4	43	<5	7	<3	10	<10	<2	<0.1	65	140	67	<5
AB92 100+00N 117+00E	Soil	<5	<0.1	220	<2	49	<5	5	<3	7	<10	<2	<0.1	31	62	44	6
AB92 100+00N 118+00E	Soil	<5	<0.1	601	6	62	<5	<5	<3	18	<10	<2	<0.1	53	74	43	<5
AB92 100+00N 119+00E	Soil	<5	<0.1	212	<2	55	<5	<5	<3	6	<10	<2	<0.1	17	64	51	<5
AB92 100+00N 120+00E	Soil	<5	0.1	883	3	69	<5	<5	<3	11	<10	<2	<0.1	28	56	137	<5
AB92 100+00N 121+00E	Soil	<5	<0.1	243	3	58	<5	<5	<3	4	<10	<2	<0.1	31	66	131	<5
AB92 100+00N 122+00E	Soil	6	<0.1	142	2	55	7	<5	<3	6	<10	<2	<0.1	24	67	88	<5
AB92 100+00N 123+00E	Soil	560	0.2	201	4	69	<5	<5	<3	4	<10	<2	<0.1	60	111	121	<5
AB92 102+00N 108+00E	Soil	<5	0.3	35	90	39	<5	11	<3	5	<10	<2	<0.1	30	86	44	<5
AB92 102+00N 109+00E	Soil	<5	0.1	219	24	85	<5	5	<3	3	<10	<2	<0.1	32	64	49	<5
AB92 102+00N 110+00E	Soil	<5	0.5	60	98	36	<5	9	<3	2	<10	<2	0.1	32	73	27	<5
AB92 102+00N 111+00E	Soil	<5	0.7	790	7	47	<5	12	<3	8	<10	<2	<0.1	33	104	6	<5
AB92 102+00N 112+00E	Soil	<5	0.3	617	3	18	<5	5	<3	12	<10	<2	<0.1	56	92	58	<5
AB92 102+00N 113+00E	Soil	6	0.4	623	3	25	<5	6	<3	12	<10	<2	<0.1	68	110	83	<5
AB92 102+00N 114+00E	Soil	<5	0.4	847	2	18	<5	7	3	7	<10	<2	<0.1	37	77	48	<5
AB92 102+00N 115+00E	Soil	10	<0.1	231	<2	27	<5	8	3	8	<10	<2	<0.1	44	79	42	<5
AB92 102+00N 116+00E	Soil	6	<0.1	254	<2	40	<5	5	<3	8	<10	<2	<0.1	30	71	57	<5
AB92 102+00N 117+00E	Soil	26	<0.1	395	6	54	<5	<5	<3	11	<10	<2	<0.1	27	61	80	<5
AB92 102+00N 118+00E	Soil	<5	<0.1	199	4	71	<5	<5	<3	9	<10	<2	<0.1	25	53	120	<5
AB92 102+00N 119+00E	Soil	6	<0.1	140	4	69	5	<5	<3	6	<10	<2	<0.1	20	48	140	<5
AB92 102+00N 120+00E	Soil	18	0.1	129	<2	56	6	5	<3	3	<10	<2	<0.1	19	61	76	<5
AB92 102+00N 121+00E	Soil	66	0.2	83	<2	67	<5	<5	<3	3	<10	<2	<0.1	27	56	98	<5
AB92 102+00N 122+00E	Soil	70	<0.1	184	<2	60	7	6	<3	2	<10	<2	<0.1	34	71	56	<5
AB92 102+00N 123+00E	Soil	436	0.1	171	2	74	<5	5	<3	4	<10	<2	<0.1	37	87	72	<5
AB92 104+00N 95+00E	Soil	16	<0.1	85	<2	27	<5	<5	<3	1	<10	<2	<0.1	18	21	26	<5
AB92 104+00N 96+00E	Soil	<5	<0.1	257	2	42	<5	5	<3	2	<10	<2	<0.1	31	43	74	<5
AB92 104+00N 97+00E	Soil	<5	0.1	146	<2	27	<5	<5	<3	1	<10	<2	<0.1	22	29	74	<5
AB92 104+00N 98+00E	Soil	<5	<0.1	221	<2	55	<5	<5	3	3	<10	<2	<0.1	32	25	165	6
Minimum Detection		5	0.1	1	2	1	5	5	3	1	10	2	0.1	1	1	2	5
Maximum Detection		10000	100.0	20000	20000	20000	10000	1000	10000	1000	10000	10000.0	10000	10000	10000	10000	10000
Method		F/AAS	ICP	ICP													

--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est

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Sample Name	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti %	Al %	Ca %	Fe %	Mg %	K %	Na %	P %
AB92 100+00N 106+00E	31	257	1081	<2	165	1	6	0.16	2.65	1.12	7.0%	3.27	0.78	0.02	0.07
AB92 100+00N 107+00E	710	118	965	<2	24	2	9	0.14	2.56	1.62	4.53	5.33	1.17	0.02	0.05
AB92 100+00N 108+00E A	584	90	883	<2	11	1	6	0.08	1.70	0.74	5.1%	3.63	0.32	0.02	0.04
AB92 100+00N 108+00E B	611	90	871	<2	11	1	6	0.09	1.98	0.78	4.73	4.25	0.33	0.01	0.04
AB92 100+00N 109+00E	409	82	663	<2	15	1	5	0.09	1.80	0.74	4.16	3.78	0.23	0.02	0.03
AB92 100+00N 110+00E	521	64	400	<2	8	1	3	0.06	0.95	0.43	4.68	2.30	0.19	0.02	0.02
AB92 100+00N 111+00E	70	179	546	<2	360	1	4	0.30	3.44	0.63	5.7%	3.88	0.30	0.03	0.01
AB92 100+00N 112+00E	369	84	516	<2	13	1	5	0.08	1.52	0.75	4.57	3.19	0.44	0.02	0.02
AB92 100+00N 113+00E	316	80	434	<2	10	1	4	0.08	1.15	0.80	4.06	2.23	0.33	0.02	0.02
AB92 100+00N 114+00E	302	55	396	<2	2	1	4	0.05	1.12	0.42	3.30	2.51	0.06	0.02	<0.01
AB92 100+00N 115+00E	462	94	563	<2	9	1	4	0.06	1.30	0.36	5.5%	2.51	0.05	0.02	0.02
AB92 100+00N 116+00E	338	114	494	<2	14	1	8	0.16	1.87	0.46	7.2%	3.06	0.17	0.03	0.02
AB92 100+00N 117+00E	280	105	709	<2	20	1	3	0.12	1.81	0.30	4.19	1.91	0.04	0.02	0.03
AB92 100+00N 118+00E	179	116	543	<2	26	1	6	0.11	1.80	0.30	4.74	1.96	0.13	0.02	0.03
AB92 100+00N 119+00E	200	170	441	<2	29	1	2	0.12	2.51	0.21	5.5%	1.57	0.14	0.02	0.07
AB92 100+00N 120+00E	127	182	737	4	33	1	5	0.11	2.54	0.69	7.1%	2.06	0.51	0.02	0.06
AB92 100+00N 121+00E	149	174	1358	<2	31	1	1	0.09	2.49	0.27	5.5%	1.78	0.35	0.02	0.11
AB92 100+00N 122+00E	163	134	513	<2	49	1	1	0.05	2.36	0.28	5.5%	1.79	0.26	0.02	0.15
AB92 100+00N 123+00E	251	230	2213	3	98	2	13	0.11	3.72	0.30	8.1%	4.29	0.38	0.02	0.14
AB92 102+00N 108+00E	389	85	503	<2	12	1	4	0.07	1.31	0.51	4.54	2.65	0.28	0.02	0.03
AB92 102+00N 109+00E	256	186	625	<2	9	1	5	0.22	2.54	0.24	6.0%	3.12	0.19	0.04	0.02
AB92 102+00N 110+00E	297	78	549	<2	4	1	5	0.09	1.45	0.37	3.36	2.64	0.17	0.02	0.01
AB92 102+00N 111+00E	438	96	343	<2	4	1	3	0.06	0.99	0.28	5.6%	2.19	0.03	0.02	0.02
AB92 102+00N 112+00E	249	66	328	<2	16	1	4	0.06	1.39	0.29	4.30	2.20	0.33	0.02	0.01
AB92 102+00N 113+00E	245	69	433	<2	16	1	5	0.07	1.77	0.37	4.53	2.90	0.42	0.02	0.02
AB92 102+00N 114+00E	279	78	285	<2	20	1	5	0.06	1.23	0.29	3.85	1.91	0.17	0.02	0.01
AB92 102+00N 115+00E	308	78	493	<2	14	<1	5	0.06	1.30	0.33	4.26	2.13	0.05	0.02	0.02
AB92 102+00N 116+00E	219	136	447	<2	56	1	2	0.06	1.84	0.43	5.3%	1.88	0.07	0.02	0.04
AB92 102+00N 117+00E	135	114	649	2	83	1	2	0.08	2.32	0.29	4.28	1.57	0.10	0.02	0.12
AB92 102+00N 118+00E	148	153	938	<2	50	1	1	0.10	2.31	0.22	4.51	1.52	0.20	0.02	0.09
AB92 102+00N 119+00E	117	137	989	2	49	1	1	0.07	2.03	0.21	4.68	1.29	0.12	0.02	0.15
AB92 102+00N 120+00E	130	125	646	<2	33	1	1	0.05	2.34	0.20	4.47	1.54	0.09	0.02	0.10
AB92 102+00N 121+00E	195	139	1335	<2	36	<1	1	0.05	2.27	0.23	6.0%	1.65	0.12	0.02	0.12
AB92 102+00N 122+00E	165	121	865	2	54	1	4	0.08	2.44	0.33	5.5%	2.23	0.24	0.02	0.13
AB92 102+00N 123+00E	239	238	2149	?	14	1	9	0.12	3.44	0.17	7.8%	3.43	0.09	0.01	0.08
AB92 104+00N 95+00E	65	60	535	<2	68	1	3	0.01	1.41	0.55	2.84	1.32	0.02	0.02	0.08
AB92 104+00N 96+00E	171	90	977	<2	88	1	6	0.08	2.39	0.47	3.50	2.56	0.12	0.02	0.07
AB92 104+00N 97+00E	117	64	424	2	73	1	2	0.04	1.65	0.33	3.07	1.19	0.04	0.02	0.06
AB92 104+00N 98+00E	54	133	1230	<2	81	<1	3	0.08	2.82	0.56	4.46	2.13	0.26	0.03	0.10
Minimum Detection	1	2	1	2	1	1	1	0.01	0.01	0.01	0.01	0.01	0.01	0.01	0.01
Maximum Detection	10000	10000	10000	10000	10000	10000	10000	1.00	5.00	10.00	5.00	10.00	10.00	5.00	5.00
Method	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP	ICP
--No Test	ReC=ReCheck	ins=Insufficient	Sample	m=Est/1000	%=Est	%	Max=No	Est							

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Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
AB92 104+00N 99+00E	Soil	<5	<0.1	195	<2	30	<5	5	<3	2	<10	<2	<0.1	26	30	73	<5
AB92 104+00N 100+00E	Soil	<5	0.1	55	<2	47	<5	<5	<3	3	<10	<2	<0.1	17	29	103	<5
AB92 104+00N 101+00E	Soil	<5	0.1	50	2	36	<5	<5	<3	3	<10	<2	<0.1	14	19	54	<5
AB92 104+00N 102+00E	Soil	<5	<0.1	89	<2	49	<5	<5	<3	4	<10	<2	<0.1	22	32	105	<5
AB92 104+00N 103+00E	Soil	<5	<0.1	168	<2	37	<5	<5	<3	2	<10	<2	<0.1	31	37	106	<5
AB92 104+00N 104+00E	Soil	<5	<0.1	109	<2	42	<5	<5	<3	2	<10	<2	<0.1	25	30	99	<5
AB92 104+00N 105+00E	Soil	<5	0.4	30	<2	26	<5	<5	<3	3	<10	<2	<0.1	15	23	67	<5
AB92 104+00N 106+00E	Soil	<5	0.5	328	190	119	<5	13	<3	17	<10	<2	<0.1	83	198	44	<5
AB92 104+00N 107+00E	Soil	<5	0.1	121	187	121	<5	13	4	13	<10	<2	0.1	58	143	57	<5
AB92 104+00N 108+00E	Soil	<5	0.3	31	92	57	<5	11	<3	4	<10	<2	<0.1	45	87	60	<5
AB92 104+00N 109+00E	Soil	<5	<0.1	32	29	47	<5	13	<3	5	<10	<2	<0.1	40	101	24	<5
AB92 104+00N 110+00E	Soil	<5	<0.1	312	11	24	<5	5	<3	20	<10	<2	<0.1	40	85	51	<5
AB92 104+00N 111+00E	Soil	<5	<0.1	492	9	25	<5	5	<3	18	<10	<2	<0.1	51	94	66	<5
AB92 104+00N 112+00E	Soil	<5	0.3	439	35	22	<5	7	<3	29	<10	<2	<0.1	54	86	58	<5
AB92 104+00N 113+00E	Soil	<5	0.2	800	2	27	<5	5	<3	17	<10	<2	<0.1	68	106	50	<5
AB92 104+00N 114+00E	Soil	<5	0.4	1207	6	44	5	7	3	9	<10	<2	<0.1	77	119	111	<5
AB92 104+00N 115+00E	Soil	<5	<0.1	404	4	65	<5	5	<3	8	<10	<2	<0.1	44	98	125	<5
AB92 104+00N 116+00E	Soil	<5	0.1	755	4	58	<5	5	<3	7	<10	<2	<0.1	38	95	66	<5
AB92 104+00N 117+00E	Soil	6	2.1	871	6	70	6	<5	<3	16	<10	<2	<0.1	60	132	111	<5
AB92 104+00N 118+00E	Soil	<5	0.2	939	5	69	<5	<5	<3	25	<10	<2	<0.1	36	126	123	<5
AB92 104+00N 119+00E	Soil	16	0.4	355	148	167	<5	<5	<3	9	<10	<2	<0.1	116	113	102	<5
AB92 104+00N 120+00E	Soil	<5	0.1	205	<2	65	5	<5	<3	3	<10	<2	<0.1	49	93	82	<5
AB92 104+00N 121+00E	Soil	22	0.1	214	3	62	<5	5	<3	3	<10	<2	<0.1	34	89	78	<5
AB92 104+00N 122+00E	Soil	26	<0.1	175	<2	79	<5	7	4	3	<10	<2	<0.1	48	88	94	<5
AB92 104+00N 123+00E	Soil	204	0.4	365	4	147	<5	8	<3	5	<10	<2	<0.1	64	99	157	<5
AB92 106+00N 108+00E	Soil	<5	<0.1	90	20	38	<5	7	3	9	<10	<2	<0.1	32	77	19	<5
AB92 106+00N 109+00E	Soil	<5	0.1	300	11	48	<5	5	<3	24	<10	<2	<0.1	50	100	60	<5
AB92 106+00N 110+00E	Soil	<5	<0.1	182	<2	38	<5	5	<3	11	<10	<2	<0.1	34	73	53	<5
AB92 106+00N 111+00E	Soil	<5	0.1	302	<2	35	<5	<5	<3	20	<10	<2	<0.1	39	69	57	<5
AB92 106+00N 112+00E	Soil	6	<0.1	207	<2	28	<5	9	<3	10	<10	<2	<0.1	35	83	7	<5
AB92 106+00N 113+00E	Soil	<5	<0.1	315	<2	28	<5	9	<3	4	<10	<2	<0.1	31	89	27	<5
AB92 106+00N 114+00E	Soil	74	0.3	945	3	40	8	<5	<3	11	<10	<2	<0.1	62	108	75	<5
AB92 106+00N 115+00E	Soil	22	0.5	1197	3	47	<5	5	<3	40	<10	<2	<0.1	70	143	62	<5
AB92 106+00N 116+00E	Soil	20	0.6	1075	<2	48	<5	<5	<3	33	<10	<2	<0.1	46	112	62	<5
AB92 106+00N 117+00E	Soil	14	0.5	408	5	33	<5	7	<3	17	<10	<2	<0.1	29	78	105	<5
AB92 106+00N 118+00E	Soil	8	0.3	408	<2	50	6	7	3	12	<10	<2	<0.1	26	84	111	<5
AB92 106+00N 119+00E	Soil	10	0.3	363	<2	51	7	7	<3	5	<10	<2	<0.1	41	140	107	<5
AB92 106+00N 120+00E	Soil	166	0.3	439	<2	77	6	8	<3	5	<10	<2	<0.1	65	108	64	<5
AB92 106+00N 121+00E	Soil	370	0.3	441	<2	72	<5	<5	<3	6	<10	<2	<0.1	56	116	54	<5

Minimum Detection: 5 0.1 1 2 1 5 3 1 10 2 0.1 1 1 2 5
 Maximum Detection: 10000 100.0 20000 20000 10000 1000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000 10000
 Method: FA/AAS ICP
 ---=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Es

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Sample Name	Type	Au ppb	Ag ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Sb ppm	Hg ppm	Mo ppm	Tl ppm	Bi ppm	Cd ppm	Co ppm	Ni ppm	Ba ppm	W ppm
AB92 106+00N 122+00E	Soil	250	0.5	489	<2	87	<5	7	3	7	<10	<2	<0.1	101	181	106	<5
AB92 106+00N 123+00E	Soil	414	0.5	273	10	95	<5	<5	3	5	<10	<2	<0.1	85	113	172	<5

Minimum Detection 5 0.1 1 2 1 5 5 3 1 10 2 0.1 1 1 2 5
 Maximum Detection 10000 100.0 20000 20000 20000 10000 1000 10000 1000 1000 10000 10000 10000 10000 10000 10000 10000 10000
 Method FA/AAS ICP
 --No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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Sample Name	Cr ppm	V ppm	Mn ppm	La ppm	Sr ppm	Zr ppm	Sc ppm	Ti %	Al %	Ca %	Fe %	Mg %	K %	Na %	P %
AB92 106+00N 122+00E	385	245	3034	3	26	2	18	0.11	3.98	0.71	10%	5.83	0.37	0.01	0.15
AB92 106+00N 123+00E	252	213	3349	<2	13	2	21	0.09	3.58	0.25	9.4%	4.11	0.26	0.01	0.13

Minimum Detection 1 2 1 2 1 1 1 0.01 0.01 0.01 0.01 0.01 0.01 0.01 0.01
Maximum Detection 10000 10000 10000 10000 10000 10000 10000 1.00 5.00 10.00 5.00 10.00 10.00 5.00 5.00
Method ICP
--=No Test ReC=ReCheck ins=Insufficient Sample m=Est/1000 %=Est % Max=No Est



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Method of Gold analysis by Fire Assay / AAS

- (a) 20.0 to 30.0 grams of sample is mixed with a combination of fluxes in a fusion pot. The sample is then fused at high temperature to form a lead "button".
- (b) The precious metals are extracted by cupellation. Any Silver is dissolved by nitric acid and decanted. The gold bead is then dissolved in boiling concentrated aqua regia solution heated by a hot water bath.
- (c) The gold in solution is determined with an Atomic Absorption Spectrometer. The gold value, in parts per billion, is calculated by comparision with a set of known gold standards.

QUALITY CONTROL

Every fusion of 24 pots contains 22 samples, one internal standard or blank, and a random reweigh of one of the samples. Samples with anomalous gold values greater than 500 ppb are automatically checked by Fire Assay/AA methods. Samples with gold values greater than 10000 ppb are automatically checked by Fire Assay/Gravimetric methods.



INTERNATIONAL PLASMA LABORATORY LTD

2036 Columbia Street
Vancouver, B.C.
Canada V5Y 3E1
Phone (604) 879-7878
Fax (604) 879-7898

Method of ICP Multi-element Analyses

- (a) 0.50 grams of sample is digested with diluted aqua regia solution by heating in a hot water bath for 90 minutes, then cooled, bulked up to a fixed volume with demineralized water, and thoroughly mixed.
 - (b) The specific elements are determined using an Inductively Coupled Argon Plasma spectrophotometer. All elements are corrected for inter-element interference. All data are subsequently stored onto computer diskette.
- * Aqua regia leaching is partial for Al, Ba, Ca, Cr, Fe, K, Mg, Mn, Na, P, Sn, Sr and W.

QUALITY CONTROL

The machine is calibrated using six known standards and a blank. Another blank, which was digested with the samples, and a standard are tested before any samples to confirm the calibration. A maximum of 20 samples are analysed, and then a standard, also digested with the samples, is run. A known standard with characteristics best matching the samples is chosen and tested. Another 20 samples are analysed, with the last one being a random reweigh of one of the samples. The standard used at the beginning is rerun. This procedure is repeated for all of the samples.

APPENDIX C

Statistical Analysis

SOIL SAMPLE GEOCHEMISTRY
ON THE ABE PROPERTY

By

A.M.S.Clark, Ph.D., FGAC, P.Geo.(B.C.)
SEGURO CONSULTING INC.

28 September 1992

INTRODUCTION

An investigation of the distribution of gold and copper in soil samples from the Abe Property was carried out between 15 August and 28 September 1992.

This report is based on an evaluation of the geochemical analyses only, the author has not visited the property.

A total of 158 samples were collected from one grid on the property in the summer of 1992.

DISCUSSION

Summary statistics and correlation coefficients have been calculated for the main elements of interest, and histograms plotted for gold and copper.

Gold:

Gold values are predominantly low, with a few high values ranging up to 560 ppb Au (see Summary Statistics Table). Gold shows no significant correlation with other elements (see Correlation Coefficient Table). The strongest correlation with

elements of exploration interest is only 0.257 (very weak correlation) between gold and zinc.

The detailed histogram of gold shows an irregular distribution that may be a normal Gaussian distribution incompletely defined due to an insufficient data set. This irregular data distribution ends at about 30 ppb, with isolated higher values of gold up to 560 ppb (see histograms).

The 'breakpoints' for the symbol sizes used on the symbol maps were determined by inspection of the detailed histogram. All 'breakpoint' values are low compared to the usual soil values in British Columbia, for which usually only values of greater than 25 ppb Au are considered of exploration significance. The following are the 'breakpoints' chosen as showing the most useful pattern of values on the maps:

Gold:	Low values	≥ 6 and < 18 ppb Au
	Medium values	≥ 18 and < 30 ppb Au
	Higher values	≥ 30 ppb Au.

The higher values are located mainly in the northeast of the grid and also to the west of the longest sample line. Other higher values occur scattered over the grid.

Copper:

Copper shows a strong correlation with molybdenum, a moderate correlation with silver, a very weak correlation with nickel and a very weak negative correlation with antimony. Copper shows medium to high background values (above 100 ppm Cu, see Statistics Table and histograms) and an irregular normal distribution ending at about 300 ppm, with a possible second population from 300 ppm to about 600 ppm. Copper high values are distributed in the northeast part of the grid, but appear to have the medium values in the higher gold area, and the higher copper values as a 'halo' adjacent to the higher gold area. The possible second population of copper values defined by the medium group (mid-sized triangles on the map) could be real as a second population of copper values genetically associated with the higher gold values, whereas the higher copper values could be copper with no genetic gold association.

Copper:	Low values	≥ 150 and < 300 ppm Cu
	Medium values	≥ 300 and < 600 ppm Cu
	Higher values	≥ 600 ppm Cu.

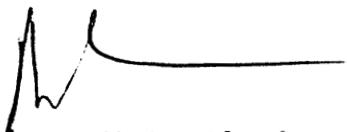
CONCLUSION

There appears to be a significant area of higher gold and associated medium copper results in the northeast part of the grid. In addition there appears to be a 'halo' of higher copper values, with little or no gold associated, forming a second population on the edge of the high gold area. High gold values have some copper associations on the west end of the longest sample-line of the grid.

CERTIFICATE

I, ANTHONY M.S. CLARK, of 2988 Fleet Street, Coquitlam, B.C., do hereby state that:

1. I am a graduate of the University of Cape Town, Cape Town, South Africa, with a Bachelor of Science Degree in Geology, 1963, and of Memorial University, St. John's, Newfoundland, with a Doctor of Philosophy Degree in Geology, 1974.
2. I am a Fellow in good standing with the Geological Association of Canada, and registered as a Professional Geologist with the Association of Professional Engineers, Geologists and Geophysicists of Alberta.
3. I actively pursued my career as an exploration geologist for twenty-three years from 1963 to 1986, since when I have undertaken consulting in the fields of mineral exploration and computer applications to exploration.
4. The information, opinions, and recommendations in this report are based on information obtained by other personnel who undertook the fieldwork on the property, and on published and unpublished literature. I have not visited the subject property.
5. I have no interest, direct or indirect, in the subject claims or the securities of Swannell Minerals Corporation.
6. I consent to the use of this report in Prospectus or Statement of Material Facts for the purpose of private or public financing.



Anthony M.S. Clark, Ph.D., F.G.A.C. P.Geo. (B.C.)

Dated at Coquitlam, B.C., 28 Sept 92

ABE Property:

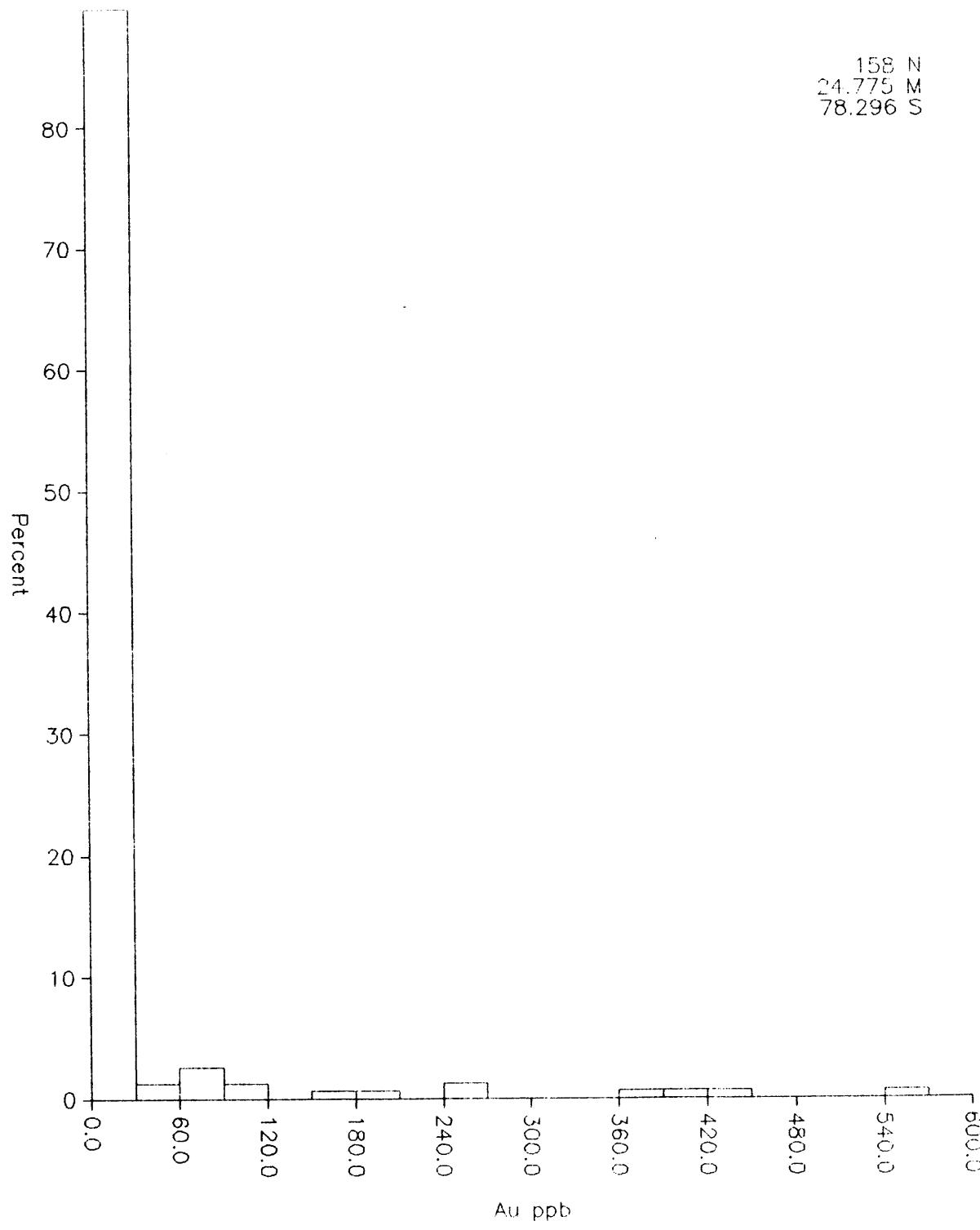
Pearson Correlation Coefficients

	Ag_ppm	As_ppm	Au_ppb	Cu_ppm	Mo_ppm	Ni_ppm	Pb_ppm	Sb_ppm	Zn_ppm
Ag_ppm	1.	0.1278	0.2145	0.4416	0.4188	0.3235	0.1987	-0.0113	0.2759
As_ppm	0.1278	1.	0.0113	0.1430	0.0367	0.0120	-0.1232	-0.1616	0.1012
Au_ppb	0.2145	0.0113	1.	0.0741	0.0577	0.2127	-0.1053	-0.1622	0.2571
Cu_ppm	0.4416	0.1430	0.0741	1.	0.6095	0.2989	-0.1797	-0.2811	0.0894
Mo_ppm	0.4188	0.0367	0.0577	0.6095	1.	0.3295	-0.0214	-0.1502	0.1147
Ni_ppm	0.3235	0.0120	0.2127	0.2989	0.3295	1.	0.2037	0.5505	0.3529
Pb_ppm	0.1987	-0.1232	-0.1053	-0.1797	-0.0214	0.2037	1.	0.3653	0.2901
Sb_ppm	-0.0113	-0.1616	-0.1622	-0.2811	-0.1502	0.5505	0.3653	1.	0.0685
Zn_ppm	0.2759	0.1012	0.2571	0.0894	0.1147	0.3529	0.2901	0.0685	1.

Summary Statistics

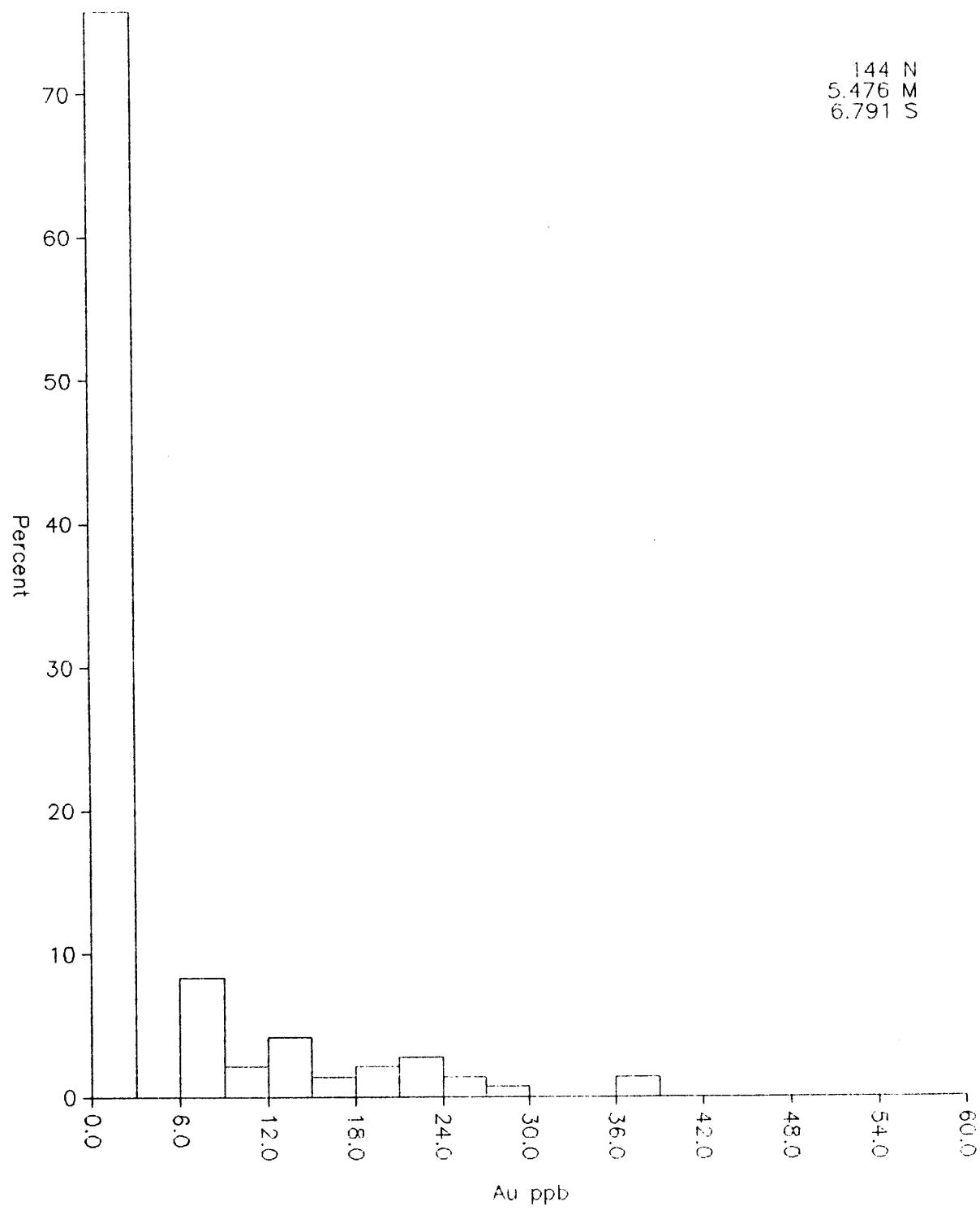
	Ag_ppm	As_ppm	Au_ppb	Cu_ppm	Mo_ppm	Ni_ppm	Pb_ppm	Sb_ppm	Zn_ppm
Number	158	158	158	158	158	158	158	158	158
Mean	0.1712	2.854	24.775	224.468	6.848	77.165	27.196	6.462	54.000
Std Dev	0.2446	1.291	78.296	268.771	7.129	34.455	49.482	3.806	25.416
Variance	0.06	1.7	6130.3	72238.0	50.6	1187.2	2448.5	14.5	646.0
Maximum	2.10	13.0	560.0	1207.0	42.0	198.0	311.0	19.0	167.0
Minimum	0.05	2.5	2.5	8.0	1.0	8.0	1.0	2.5	15.0
Range	2.05	10.5	557.5	1199.0	41.0	190.0	310.0	16.5	152.0
Coef Var	142.8968	45.2265	316.0260	119.7368	104.0966	44.6515	181.9444	58.8983	47.0672
Std Err	0.0195	0.1027	6.2289	21.3823	0.5671	2.7411	3.9366	0.3028	2.0220
Median	0.050	2.50	2.50	119.50	4.00	77.00	5.00	6.00	49.00
Mode	0.05	2.5	2.5	37.0	3.0	61.0	1.0	2.5	47.0
Skewness	4.1916	4.5943	4.6910	1.9271	2.3925	0.3901	2.9710	0.7955	1.3447
Kurtosis	25.4469	25.8990	23.0996	3.2355	6.7709	0.3277	10.1840	0.0189	2.8869

ABE: Au ppb. All Samples.



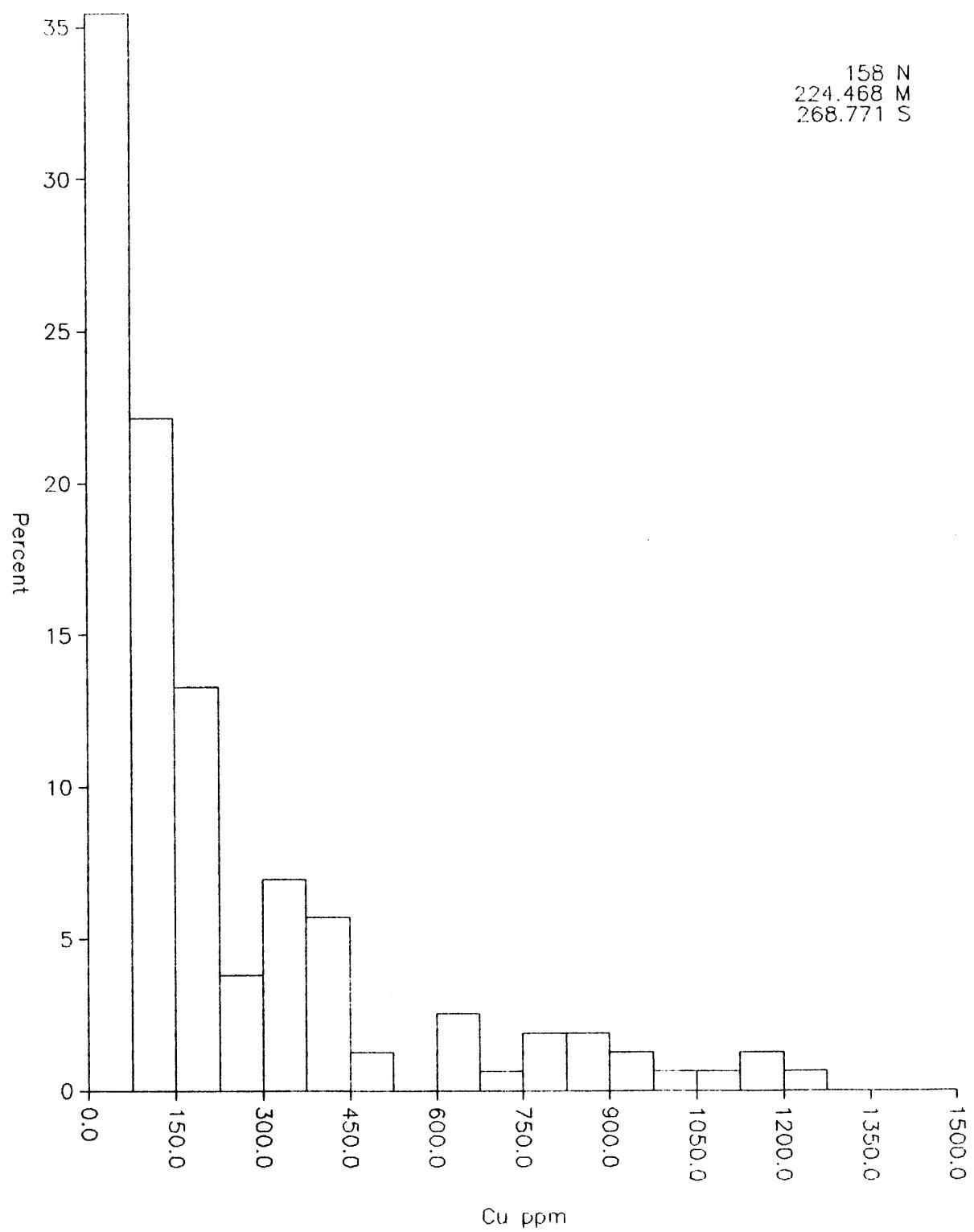
ABE: Au ppb. Partial Set.

¹⁴⁴ N
5.476 M
6.791 S



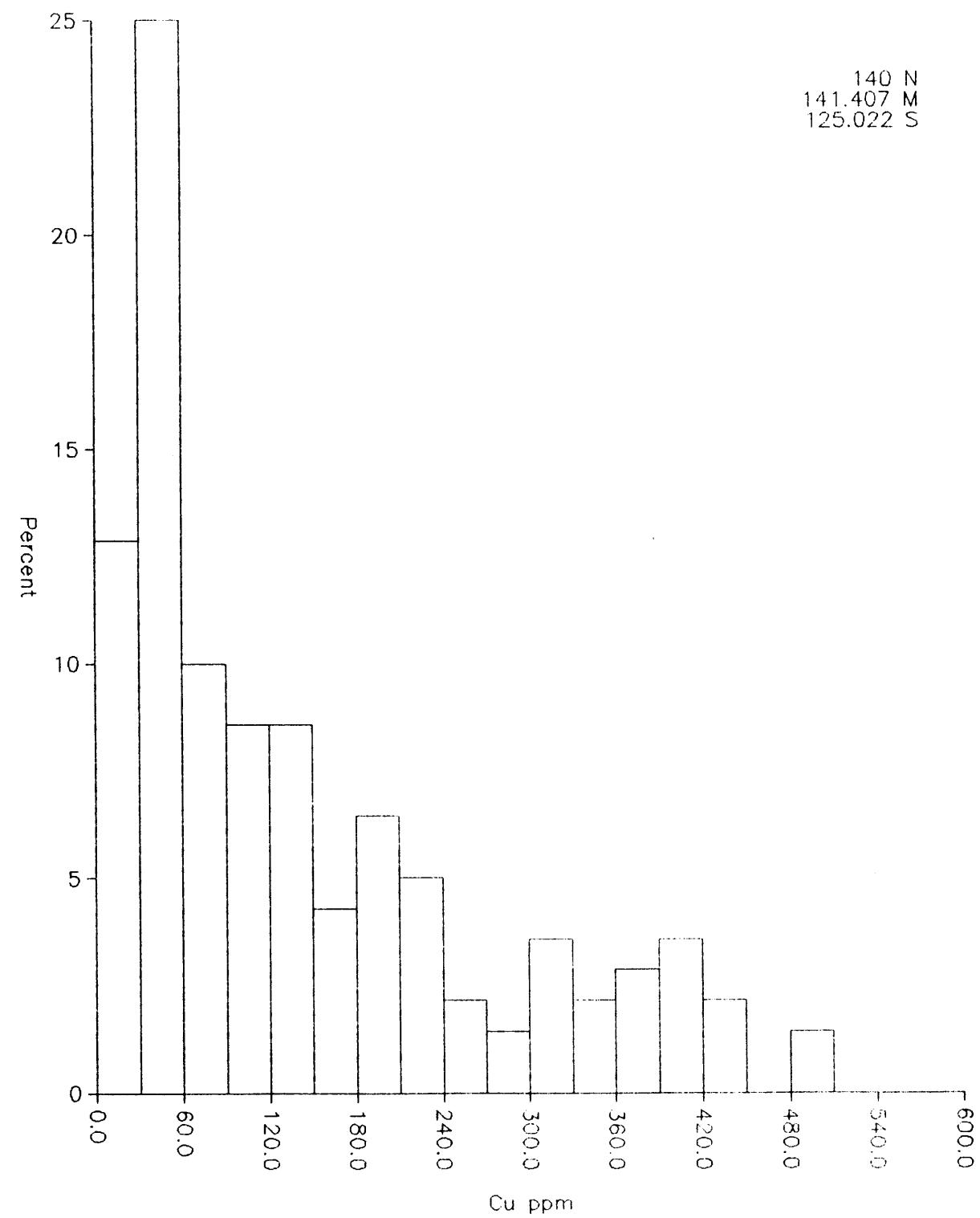
ABE: Cu ppm. All Samples.

158 N
224.468 M
268.771 S



ABE: Cu ppm. Partial Set.

140 N
141.407 M
125.022 S



APPENDIX D

PETROGRAPHIC DESCRIPTION



Vancouver Petrographics Ltd.

JAMES VINNELL, Manager
JOHN G. PAYNE, Ph.D. Geologist
CRAIG LEITCH, Ph.D. Geologist
JEFF HARRIS, Ph.D. Geologist
KEN E. NORTHCOTE, Ph.D. Geologist

P.O. BOX 39
8080 GLOVER ROAD
FORT Langley, B.C.
VOX 1J0
PHONE (604) 888-1323
FAX. (604) 888-3642

Peter D. Leriche
Reliance Geological Services Inc.
241 East 1st Street
North Vancouver, B.C. V7L 1B4
Tel. 984-3663, Fax 988-4653

Sept. 22/92
JOB #.070

Dear Peter,

Re: Petrographic Description Reliance-1

The petrographic description for the Reliance-1 sample has been completed.

I am sending a copy by Fax with hard copy, thin section and sample rejects to follow by mail.

If you wish to discuss the results call me anytime at [Telephone/Fax] 796-2068.

Yours very truly,

A handwritten signature in black ink that reads "Ken Northcote".

K.E. Northcote, Ph.D., P.Eng.

There is a total of 3 pages including this one.

Reliance.070

Crushed altered porphyritic andesite/latite andesite (K-feldspar introduced)

Summary description

Protolith porphyritic with outlines of "crushed" plagioclase and mafic phenocrysts discernable. In a crushed fine grained altered groundmass similar to phenocryst but more abundantly chloritic and with abundantly introduced(?) K-feldspar.

Plagioclase phenocrysts, crushed, outlines visible, patchy felted sericite alteration and fracture-controlled chlorite. Mafics chloritized with carbonate patches, strong cleavage control.

Groundmass strong crushed fabric composed of sericitic plagioclase and chlorite altered mafics as for phenocrysts. Crushing void networks filled with chlorite and K-feldspar (conspicuous in stained slab), lesser carbonate.

Stronger linear fractures and larger breccia voids filled with segregated chlorite, carbonate and K-feldspar.

Note: No secondary biotite, but abundantly chloritic.

Opaques; Pyrite, magnetite > chalcopyrite. Pyrite and chalcopyrite mainly fracture controlled but disseminations also occur. Magnetite diffuse clusters in chloritized mafic. Hematite (specular) associated with chalcopyrite. Few alteration rims on pyrite.

Microscopic description

Phenocrysts

Plagioclase; 15%, anhedral/subhedral (0.2 to >2.0 mm) Shattered grains, generally indistinct outlines. Patchy sericitic alteration. Shattered, fractures filled with chlorite, lesser K-feldspar.

Chlorite-altered mafic; 10%, anhedral (0.2 to 3.0 mm). Crushed, original outlines visible, pseudomorphous alteration to chlorite with small clusters of carbonate. Associated clusters of magnetite.

Groundmass: Crushed fabric

Plagioclase; 25%, anhedral (<.05 to 0.2 mm). Indistinct interlocking grains in a chlorite-rich shatter fracture network. Weak sericite slightly stronger carbonate alteration clusters.

Chlorite; ?, anhedral, (<.01 to 0.2 mm?). Altered mafics. Indistinguishable from fracture network controlled chlorite..

Crushing networks

Chlorite; 30%, anhedral (<.05 to 0.1 mm). Felted to weakly foliated blades. Fracture controlled networks among plagioclase phenocrysts and groundmass.

Reliance.070 Continued

K-feldspar; 15%, anhedral (<.05 to 0.2 mm). Most conspicuous in stained slab in groundmass. Scattered coarser grains. Observed in thin section as well defined fracture fillings.

Linear fracture fillings/veins

Chlorite; late fracture fillings cutting through groundmass and phenocrysts.

Carbonate; minor amounts associated with K-feldspar veinlets.

K-feldspar; conspicuous in thin section as minute irregular veinlets associated with lesser carbonate.

Reflected light

Opaques; about 5%

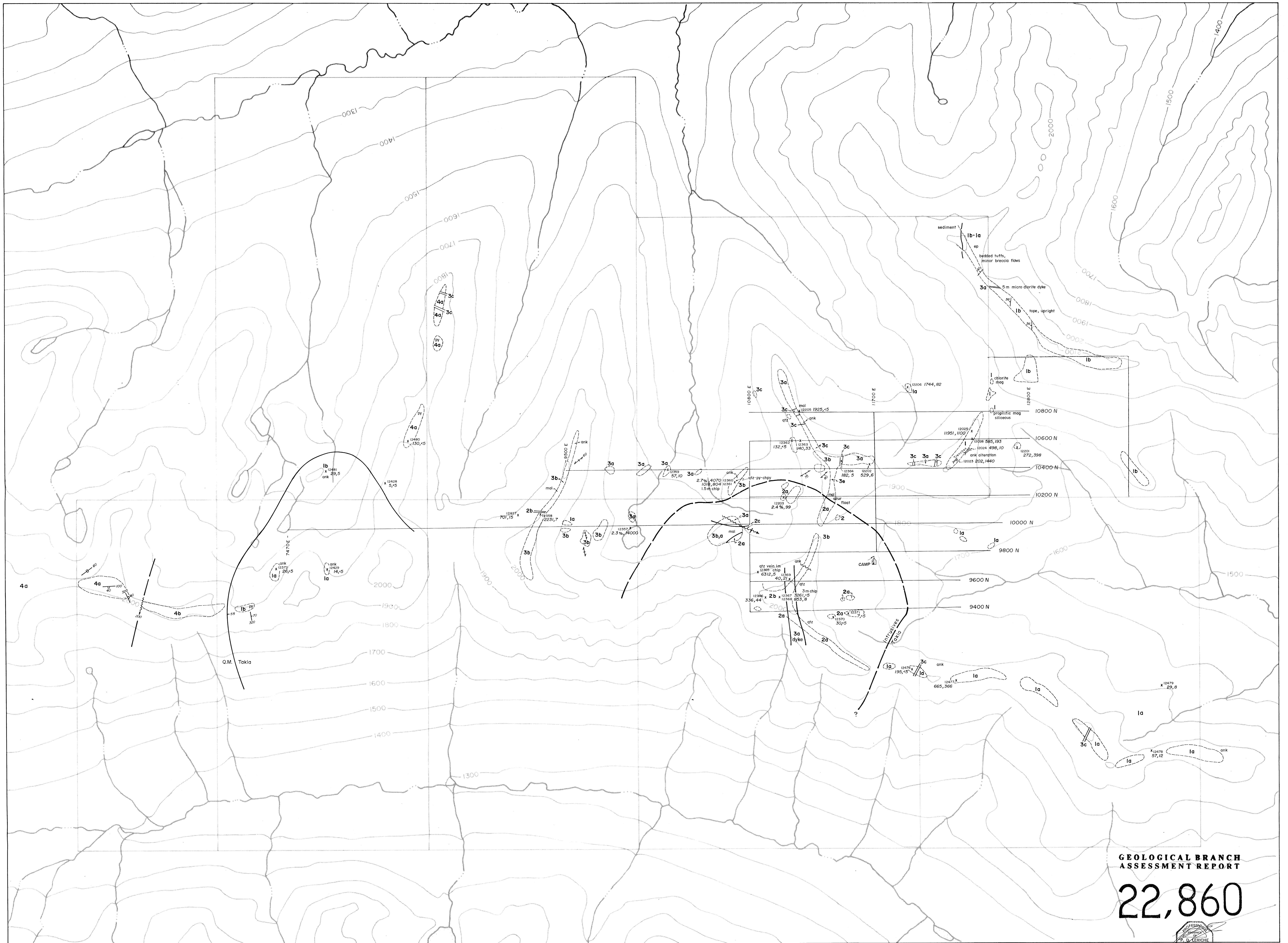
Pyrite; <2%, euhedral/anhedral (<.01 to 0.2 mm). Clusters of grains, strong fracture control. Associated chalcopyrite.

Chalcopyrite; 1%, anhedral (<.01 to >0.1 mm). Clusters of grains, some disseminated but strong fracture control. Associated interstitial to magnetite and pyrite.

Magnetite; 2%, euhedral, (<.01 to 0.4 mm, most grains <0.2 mm). Single grains, clusters of interlocking grains associated with chlorite altered mafic phenocrysts.

Hematite; <<1%, anhedral, (<.01 to <.05 mm)

- [a] specular felted clusters intergrown with chalcopyrite.
- [b] alteration rims on pyrite.



Volcanics

- I - TAKLA
 1a - andesitic augite porphyry flows and dykes
 1b - andesitic tuffs

Intrusives

- 2a - pyroxenite, massive medium grain dark green
 2b - pyroxenite chlorite rich, dark green to black
 2c - pyroxenite, gabbro phase medium grain, light green weathering
 2d - pyroxenite carbonate talc altered, moderately sheared

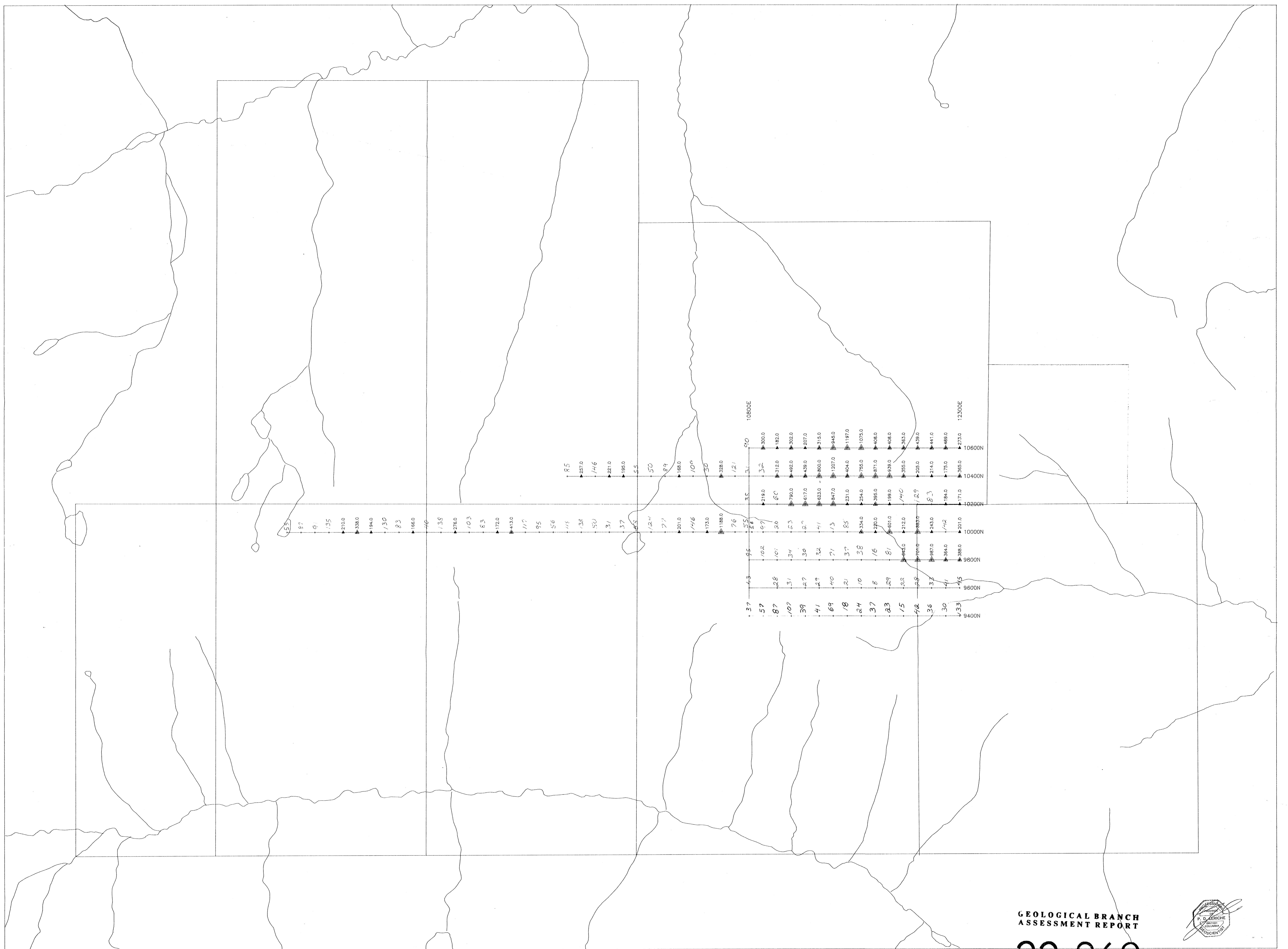
- 3a - light gray-green diorite massive and dyke-like
 3b - hybrid zone, variable amounts of angular pyroxenite xenoliths set in a diorite matrix
 3c - feldspar porphyry flows and dykes

- 4a - monzonite massive
 4b - quartz monzonite

qtz - quartz
 ank - ankerite
 hem - hematite
 mag - magnetite
 mal - malachite
 py - pyrite
 chpy - chalcocite

- outcrop
 geological contact (defined, assumed)
 shearing
 bedding
 jointing
 dyke
 rock sample location and i.d.
 $Cu (ppm)$, $Au (ppb)$

SWANNELL MINERALS CORPORATION		
ABE PROPERTY		
OMINECA M.D., B.C.		
GEOLOGY and GEOCHEMISTRY		
Scale 1:10,000	N.T.S. 94-C/5E,5W	Drawn by g.e.l.
Date August 1992	Geologist	Figure 6
REBAGLIATI GEOLOGICAL CONSULTING LTD.		
RELIANCE GEOLOGICAL SERVICES INC.		



22,860

GEOLOGICAL BRANCH
ASSESSMENT REPORT

SWANNELL MINERALS CORPORATION

FIGURE 7

ABE PROPERTY
British Columbia

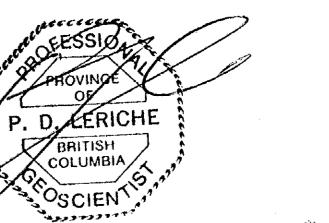
OMINECA M.D.
94C/5E,5W

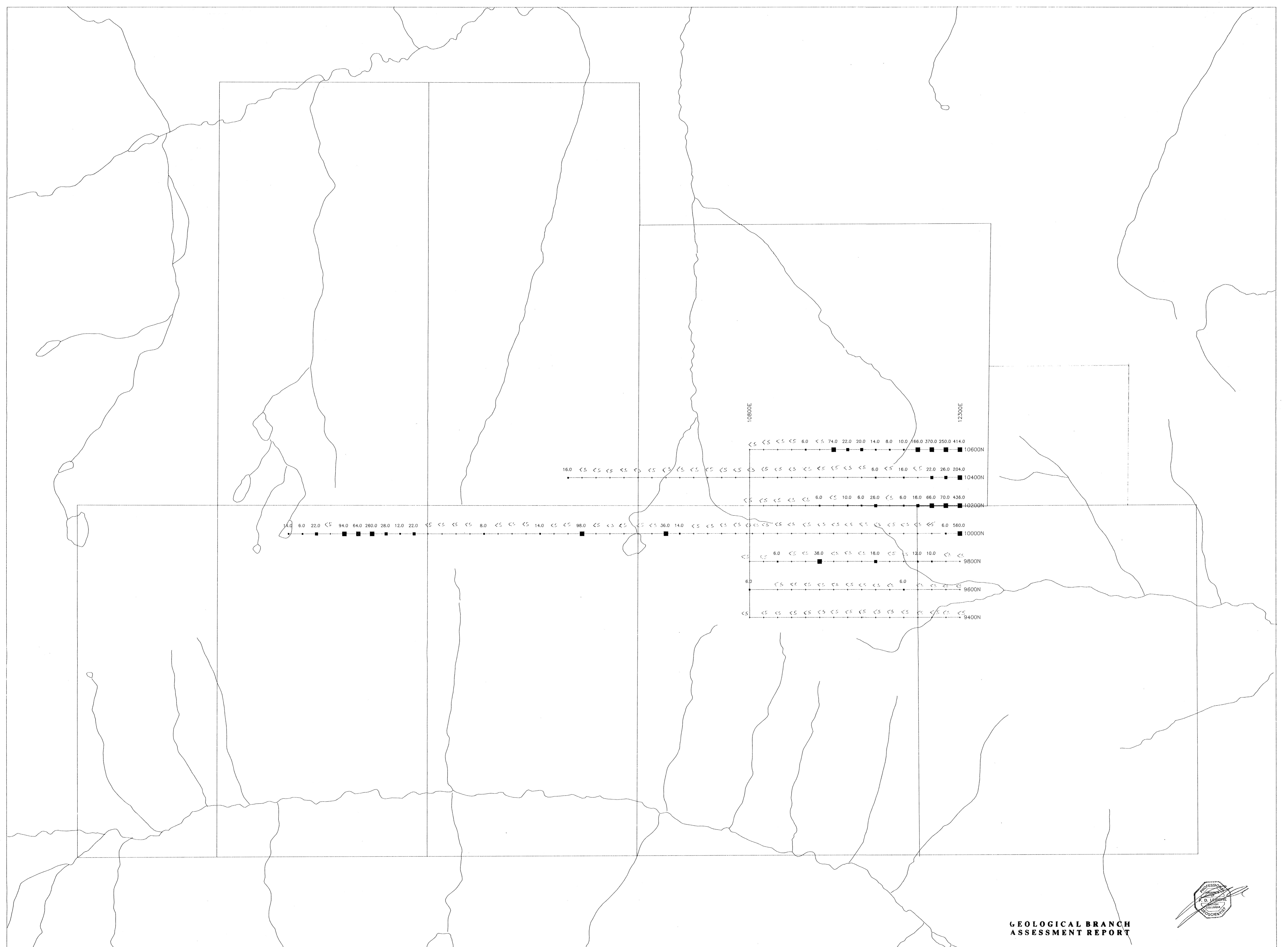
DETAILED SOIL GEOCHEMISTRY
COPPER

RELIANCE GEOLOGICAL SERVICES INC.

DATE: 24 Sept. 1992 | SCALE: 1 : 10000

Drawn By: TONY CLARK CONSULTING





22,860

SWANNELL MINERALS CORPORATION

FIGURE 8

ABE PROPERTY
British Columbia

OMINECA M.D. 94C/5E,5W

DETAILED SOIL GEOCHEMISTRY

GOLD

RELIANCE GEOLOGICAL SERVICES INC.