

84-1282-13303
11/85

REPORT ON THE
KALUM MINERAL CLAIM GROUP
FOR
BRADNER RESOURCES LTD.
SKEENA MINING DIVISION
BRITISH COLUMBIA

NTS 103I/10 and 15W
LAT. 54° 41' North, LONG. 128° 45' West

George Cavey
Diane Howe
November 19, 1984

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

13,303

SUMMARY

The Kalum mineral claim group held in agreement by Bradner Resources Ltd. of Vancouver is located along the west shore of Kitsumkalum Lake approximately 32 kilometers north of Terrace, B.C.

The oldest rocks in the claim area are argillites and greywackes belonging to the Upper Jurassic-Lower Cretaceous Age Bowser Group. Intrusive to the sediments are suites of granitoid rocks including granodiorite, diorite, monzonite and related rocks collectively termed the Coast Intrusions. Hybassal dikes believed related to the latter stages of the Coast Batholith also form a distinctive group within the claim area.

Exposed mineral showings on the Kalum property are the old "Portland" showings which consist of two separate quartz veins which contain appreciable values in gold and silver and are host in a quartz diorite stock.

The only recorded work on the Portland showing was in 1922-1923 when Kalum Mines Ltd. sunk two inclined shafts on the main vein, one to 30 feet and the other to 60 feet from which a drift was run westerly for 210 feet. Three hundred feet southeast of the shafts an adit had been driven for 85 feet to follow the second vein.

In late November 1983 and April 1984, the Kalum Lake Mining Group Ltd. did some hand trenching and blasting with follow up backhoe trenching on the area of

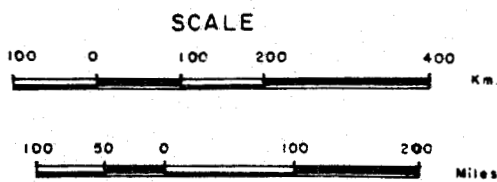
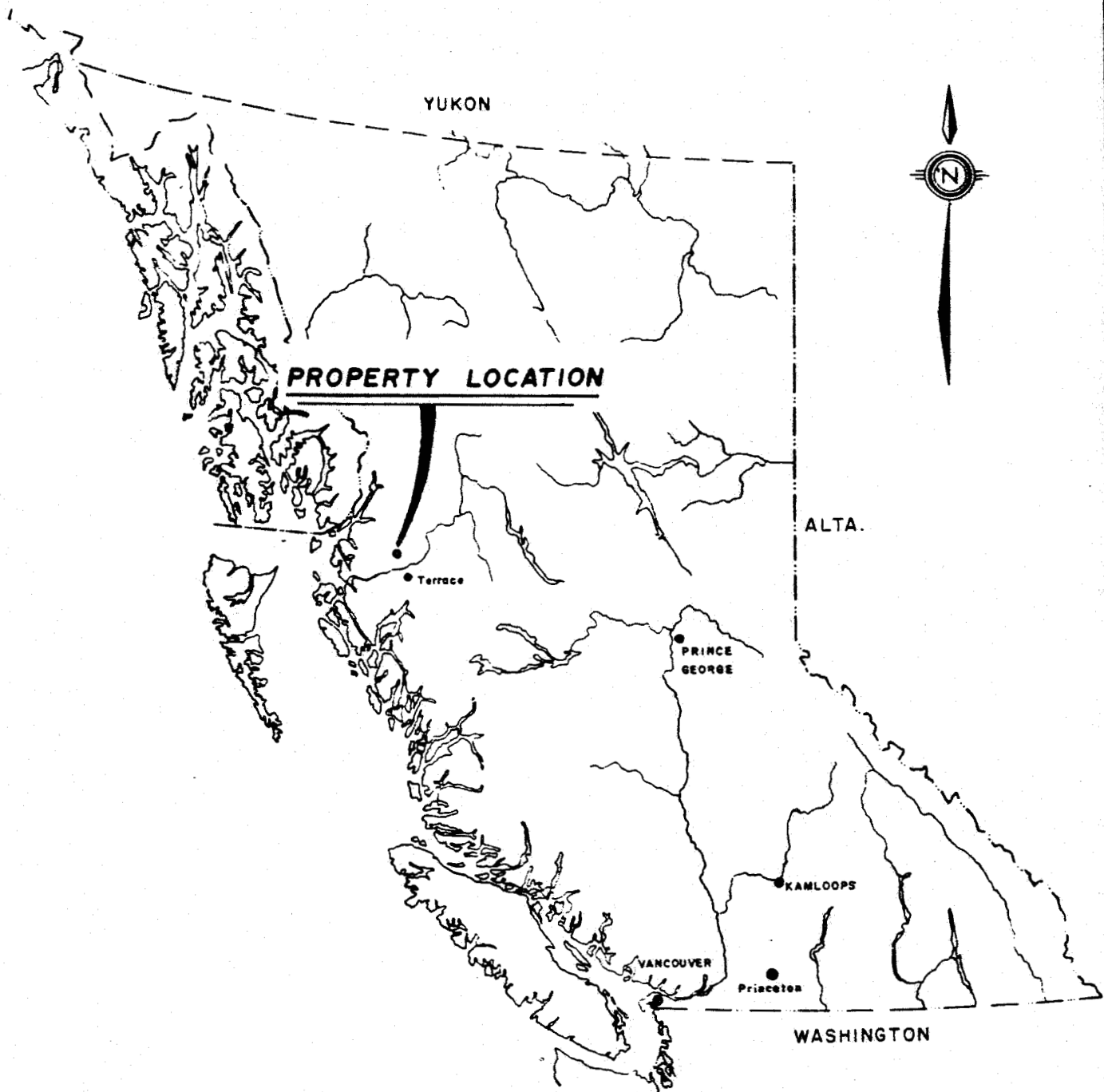
the old "Portland Group". The second vein was extended along strike for approximately 30 metres.


The Kalum Lake Mining Group also conducted some regional prospecting from which a third area of interest was discovered and subsequently staked.

The 1984 field program consisted of a soil geochemical survey over the third area (south showing) of which 576 soil and 17 rock samples were collected in an attempt to detect more auriferous quartz veins.

Results of this years program has produced favourable and encouraging results enough to recommend continuation of the work program.

Phase II should consist of linecutting (20 kilometers), detailed sampling, mapping and trenching. Phase III contingent on favourable results of Phase II should consist of 1,000 metres of overburden drilling or 375 metres of diamond drilling.



 OREQUEST CONSULTANTS LTD.

B.C. LOCATION MAP

TABLE of CONTENTS

Summary

Table of Contents

List of Figures

Appendix

1. Introduction	1
1a. Location and Access	1
1b. Physiography	2
1c. Property Information	3
1d. History and Previous Work	3
2. 1984 Field Work	5
2a. Geology	5
2b. Soil Geochemical Survey	8
3. Conclusions and Recommendations	10

Cost Estimates

Statement of Qualifications

 George Cavey, Consulting Geologist

 Diane Howe, Project Geologist

Bibliography

LIST of FIGURES

Figure 1	Property Location	following Summary
Figure 2	Claim Location	following page 3
Figure 3	Regional Geology	following page 5
Figure 4	Soil Geochemistry - Gold	In Pocket
Figure 5	Soil Geochemistry - Silver	In Pocket
Figure 6	Soil Geochemistry - Arsenic	In Pocket
Figure 7	Soil Geochemistry - Lead	In Pocket
Figure 8	Soil Geochemistry - Zinc	In Pocket
Figure 9	Soil Geochemistry - Copper	In Pocket
Figure 10	Sample Locations	

LIST of APPENDICES

Appendix 1 - Vangeochem Analysis and Techniques

Trench Plan.

1. INTRODUCTION

This report is a summary report on the soil geochemical survey conducted in late September of this year on a portion of the Kalum mineral claim group held in agreement by Bradner Resources of Vancouver.

This report briefly summarizes exploration done to date and presents recommendations for further work. Information contained in this report is from data collected during late September and early October of this year, the authors field examination in late September, as well as information obtained through various government and private publications listed in the bibliography.

1a. LOCATION and ACCESS

The Kalum mineral claim group is 32 kilometers north of the city of Terrace located in west-central B.C. The claim block is situated on the west shore and partly straddles Kitsumkalum Lake, and is centered at 54° 45' North Latitude and 128° 45' West Longitude on NTS map sheet 103I/10, 15W (Figure 1).

Easy access is provided to the claims by an all weather gravel road which leaves the Yellowhead #16 Highway approximately 5 kilometers west of Terrace and proceeds for 32 kilometers passing through the middle of the claim group.

The majority of the claim group is accessible by several old, 2 wheel drive logging roads which exit off the main access road.

Supporting infrastructure is well established with the main power transmission line which supplies power to the Nass Valley passing through the

claim group, while the CNR Prince Rupert rail line which roughly follows the Yellowhead #16 Highway across B.C. is located 32 kilometers south of the property.

Pacific Western and Canadian Pacific Airlines have daily scheduled flights from Vancouver to Terrace daily.

1b. PHYSIOGRAPHY

The property is located at the divide of the Pacific Ranges of the Coast Mountains and the Hazelton Mountains of the Intermontane Physiographic Belt.

The Kitsumkalum Valley is typical of a wide glaciated valley with flat, gently rolling valley bottoms to steep, rugged mountain flanks. Elevations on the property vary from 500-1,500 feet ASL.

The area is well timbered with cedar, hemlock, fir and spruce with choking intergrowths of alder and willow.

The majority of the claims lie on the west shore of Kitsumkalum Lake which would provide enough water for any further exploration and development. The Nelson River, which crosscuts through the Burn 2 and 3 claims, would also provide adequate water for any drilling to be done. It should be noted at this point that the majority of the Trench claims are overlain by Kitsumkalum Lake.

Thick glacial debris consisting of clay, sand and till blanket at least 60% of the claim area.

1c. PROPERTY INFORMATION

The Kalum group of claims consist of 5 claim blocks totalling 87 units.

All claims are owned by J. Apolzer of Terrace, B.C. and by agreement Bradner Resources Ltd. of Vancouver. The details of the agreement are beyond the scope of this report.

The following table summarizes pertinent data for the claim block:

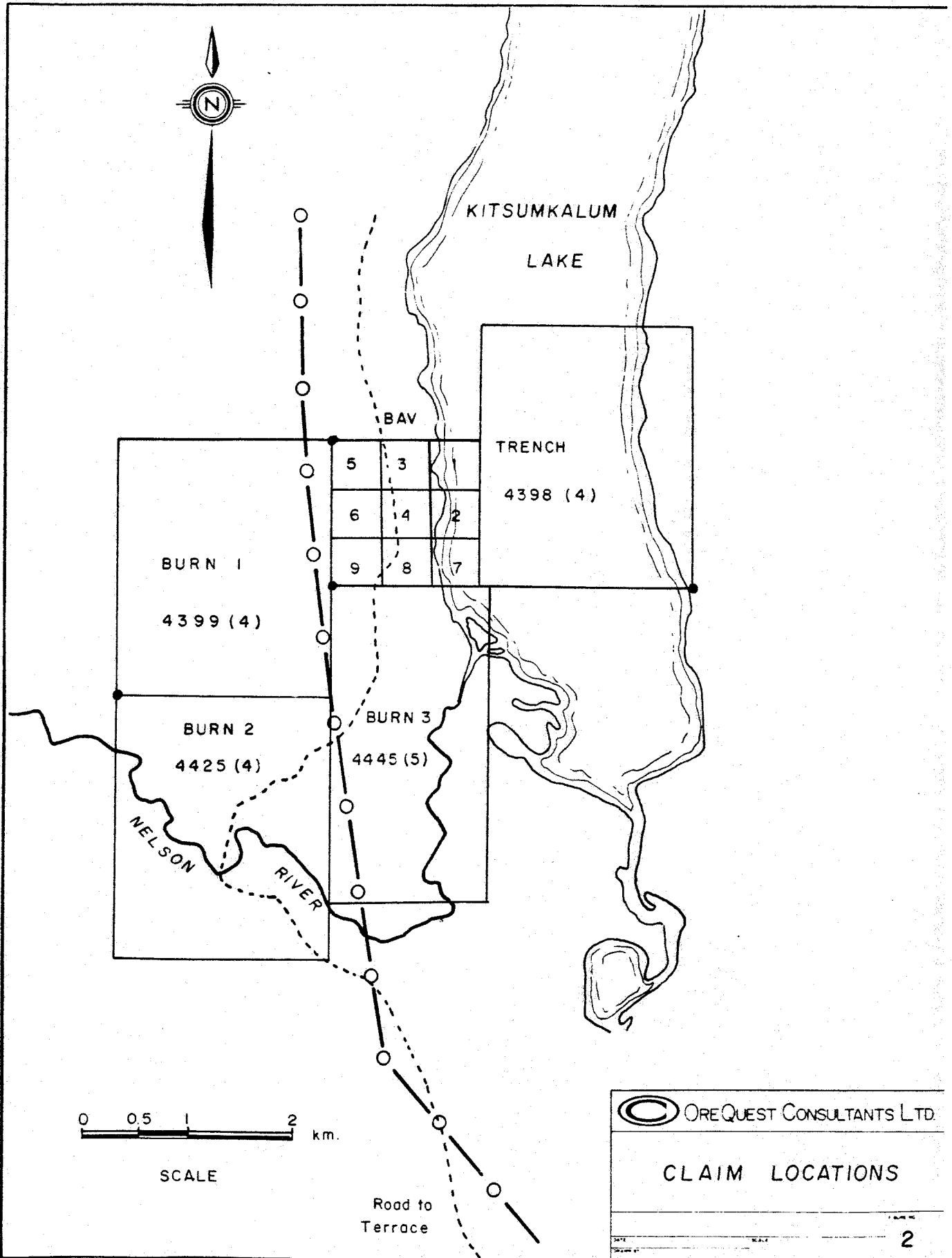
Claim Name	Units	Record #	Anniversary Date	Year*
Bav 1-4	4	37397-37400	July 21, 1990	1994
Bav 5-9	5	4223-4227	November 28, 1984	1994
Trench 1	20	4398	April 13, 1985	1988
Burn 1	20	4399	April 13, 1985	1987
Burn 2	20	4425	April 27, 1985	1987
Burn 3	18	4445	May 11, 1985	1987


*Pending approval of this years assessment.

All claims are located in the Skeena Mining Division of B.C. (see figure 2).

1d. HISTORY and PREVIOUS WORK

Earliest recorded activity on the Kalum property is 1919 when C.A. Smith of Terrace staked the original Lakeside claims, with the Portland and West Portland claims to follow in 1922. Between 1923 and 1925 the newly formed Kalum Mines Ltd. conducted considerable work on the property, which consisted of shaft sinking and drift development along the main vein discovered in 1919. Two shafts were sunk with the east shaft reaching 30 feet (9.1 metres) depth and the main or west shaft developed to 60 feet (18.2 metres) with 210 feet (64.0



 OREQUEST CONSULTANTS LTD.

CLAIM LOCATIONS

metres) of drifting westerly along the vein. A selected grab sample collected in 1930 assayed 0.62 oz/t gold and 2.2 oz/t silver.

Approximately 300 feet southeast of the main vein, Kalum Mines Ltd. put in a third adit along a #2 vein extending 85 feet (25.91 metres). Assay values from this vein in 1937 contained only minor amounts of gold and silver.

In 1972, the original claims were restaked as the Bav 1-4 by J. Apolzer of Terrace, B.C. One drill hole totalling 374 feet was drilled in an attempt to intersect the main vein. Drill records indicate that the main vein was not intersected, but granodiorite with areas of quartz veining and minor alteration were intersected. Gold and silver values range from 0.002-0.011 oz/ton and 0.08-0.02 oz/ton respectively. From studying the drill hole plan, it appears that this drill hole was drilled almost parallel to the strike of the main vein.

In November of 1983, Kalum Lake Mining Group, of which Mr. Apolzer the present property vendor is a partner, trenched and sampled the Main and #2 veins. Values up to 7.328 oz/t gold and 6.58 oz/t silver were received in a few grab samples collected from the #2 vein extension. A total of five trenches were dug utilizing a traxcavator backhoe accompanied with blasting and hand trenching. Several of the trenches did not reach bedrock and were abandoned after the stability of the slopes became hazardous.

Just to the west of the Kalum Lake Mining claims, Campbell Resources Ltd. is presently conducting a drilling program on their Misty claims. Staked as a

result of the release of the 1979 Silt Geochemistry Survey by the Government, Campbell Resources Ltd. has outlined areas of high gold interest by using soil geochemistry in what is felt to be the same type of environment as the Kalum mineral showings. No further work has been reported since 1982, but Campbell is still drilling on this property.

2. 1984 FIELD WORK

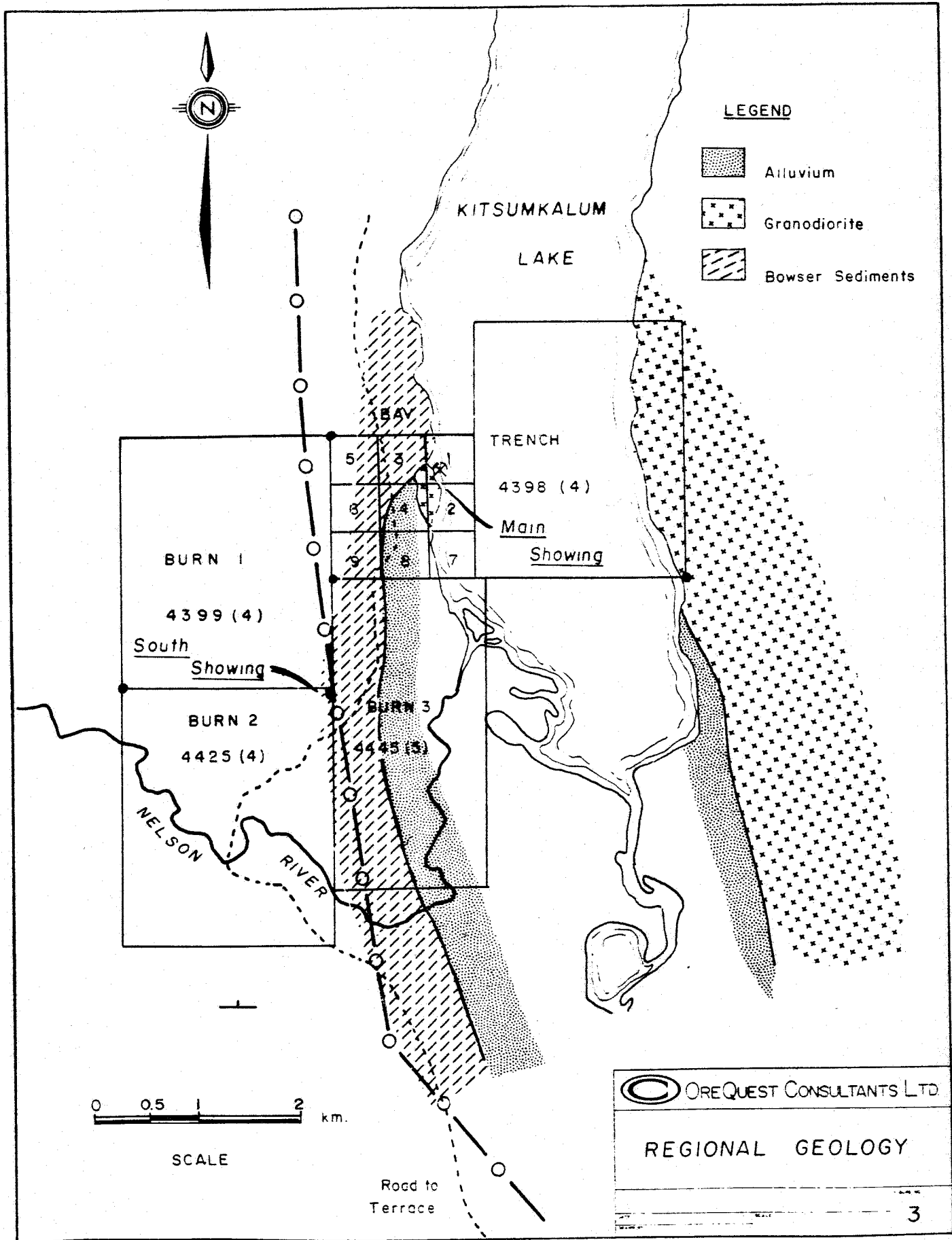
The 1984 field work was carried out by Mr. Ian Dow of Terrace under the guidance of Diane Howe, OreQuest Consultants Ltd. of Vancouver. The quality of work carried out by Mr. Dow and the efficient manner in which the work was completed has been excellent.

2a. GEOLOGY

Bedrock exposure along the valley bottom is sparse and is largely confined to the shore of Kitsumkalum Lake, streams, gulleys and old trenches on the property. A thick layer (up to 60 metres) of glacial alluvium masks at least 60% of the claim area. Towards the west edge of the Burn 1 and 2 claims outcrop exposure becomes more noticeable.

The oldest rocks underlying the claim area are Upper Jurassic-Lower Cretaceous dominantly greywackes, conglomerates and argillites belonging to the sedimentary package of Bowser Group. General strike in the claim area is east-west with dips 75° northerly. Intrusive to this sedimentary package are stocks of the Coast Intrusions which consist of granodiorite, diorite, quartz diorite and quartz monzonite of Upper Cretaceous or later age (Figure 3).

Hybassal rocks in the form of dikes and sills varying between porphyritic to



aplitic to basaltic types intrude both the sediments and Coast granitoids.

Mineralization at the Kalum property is of the epigenetic vein type which typically consists of quartz gangue, pyrite, chalcopyrite and tetrahedrite with associated values in gold and silver. Lode vein deposits are common throughout the Terrace area with most deposits consisting of narrow quartz veins that have formed in faults, fractures, shear zones or along margins of dikes.

There are two good examples of vein mineralization exposed on the Kalum property and are host in a small granodiorite stock near waters edge.

The main vein, which was the locus for work in 1922-1923, is about 30 centimetres (1 foot) true width as exposed in the two shafts. Mineralization consists of pyrite, chalcopyrite, tetrahedrite and quartz gangue with associated gold and silver values. Selected assay samples collected from the dump between 1978-1984 have assay values ranging between trace to 5.62 oz/ton gold and 0.01-13.92 oz/t silver. Both shaft were visited earlier this spring and fall and both were caved and probably flooded at depth.

The #2 vein, which is believed to be the vein which was followed by the adit in 1923, has been trenched and exposed on surface for about 30 metres (100 feet) along strike. This vein similar to the main vein in mineralogy, varies between 15 to 60 centimetres (6 to 24 inches) true width. In reports by the B.C. Ministry of Mines, there is mention of another vein approximately 10 centimetres wide which comes in along the north wall of the old adit and comes to within 5 centimetres of the second vein.

Selected assay samples taken from the adit in 1937 indicate only minor amounts of gold and silver. Surface trench samples taken from the same vein in 1983-1984 have yielded values up to 7.328 oz/ton gold and 6.58 oz/ton silver.

Both the main and #2 veins strike parallel at 037° northeast with the main vein dipping approximately 45° southwest and the #2 vein dipping approximately 65° southwest.

Not all quartz veins crosscutting the granodiorite carry mineralization. Most quartz veins observed are generally a barren, milky coloured quartz which strike oblique to the two main auriferous veins. It is felt that the masses of barren quartz are older in age than the auriferous veins, but this is uncertain at this time.

Both basaltic and porphyritic type dikes are observed within the old workings. In review of the old drill core left on the property, both types appear fairly common, but their relationships to the mineralization at this time is uncertain. A 12 cm wide basaltic dike was observed as the footwall in the main vein adit. It is possible that the open spaces developed along the borders of the dikes formed conduits or dilatant zones for mineral deposition.

Aplitic dikes are frequently observed crosscutting the barren quartz veins. The relationship again to the other rocks observed is uncertain.

A second area of interest located approximated 2.25 kilometers southwest of

the main showing has been discovered by the owners while doing some regional prospecting. Tentatively labeled the south showing, the granodiorite here is similar to the main showing and indicates some alteration to the original rock with associated quartz veins and stringers. Pyrite and chalcopyrite have been observed and selected grab samples from a trench have yielded values up to 0.49 oz/t gold and 7.06 oz/t silver.

2b. SOIL GEOCHEMICAL SURVEY

Most exploration and mining personal believe that soil geochemistry is an effective tool in areas of little or no glacial overburden. Exploration geochemistry is based on the feature that economic mineral deposits constitute an "anomalous" concentration of one or more metals relative to surrounding country host rock. During the process of surface and near surface weathering, these anomalous concentrations of metals become incorporated in the weathering products and through natural processes of chemical and mechanical breakdown spread outwards from the ore deposit giving a "dispersion halo", which provides a considerably larger exploration target than the ore deposit itself.

Based on the abovementioned theory, a systematic grid was established over the Burn 1 and 2 claims where glacial overburden was believed minimal and mineralization has been observed.

A total of 576 soil and 17 rock samples were collected along grid lines 100 metres apart with 100 metre interval stations. A four kilometer cut line base was used for control while grid lines were established using a compass and chain.

Soil samples of the B horizon were collected where possible and sent to Vangeochem Labs in Vancouver for analysis in gold, silver, copper, lead, zinc, arsenic and molybdenum. A summary of their techniques and copies of the lab report sheets are given in Appendix I.

From results returned, threshold, anomalous and very anomalous values were determined by standard statistical techniques.

	Threshold	Anomalous	Very Anomalous
Gold (ppb)	29	52	95
Silver (ppm)	1.4	2.5	4.6
Arsenic (ppm)	23	53	121
Zinc (ppm)	222	449	909
Lead (ppm)	43	70	114
Copper (ppm)	36	64	112

Results of the survey are encouraging.

Most of the anomalies in all elements are isolated occurrences, but based on the large grid dimensions the individual anomalies are expected. An area stretching between Lines 2+00S and 7+00N and stations 0+00N (baseline) to 20+00N should be considered for further detailed evaluation.

Gold values returned are extremely encouraging with isolated anomalous values reaching 9,400 ppb. Most of the anomalous values are confined between Lines 2+00S to 7+00N and stations 0+00 (baseline) to 6+00 West. This area partly overlies the south showing and is believed underlain by the altered granodiorite. A large swamp passes through the middle of the anomalous area

indicating that overburden may vary in thickness through this area.

One other very anomalous isolated gold value (120 ppb) is located outside the area of interest on Line 12+00S at Station 15+00W. This small area should also be resampled.

High silver values are primarily confined to the main area of interest. Silver values range between "0" to 3.1 ppm. The higher anomalous silver values are located on Lines 1+00S at station 1+00W, 1+00W at station 14+00W and 5+00W at station 14+00W. Elevated silver values are generally dispersed, but notably confined to the area of interest. Arsenic is the same as with silver values. Isolated arsenic anomalies are notably confined to the area of interest. There are also scattered isolated anomalies in zinc, copper and lead which either may or may not be coincident with one other element.

3. CONCLUSIONS and RECOMMENDATIONS

Phase I has been successfully completed and consisted of a soil geochemical survey and a preliminary geological survey over the Burn 1 and 2 mineral claims of the Kalum group.

Due to a thick glacial overburden cover on most of the Bav, Trench and Burn 3 claims, the entire Phase I program was designed to concentrate on areas where the overburden cover is not as excessive.

Results of the Phase I program were extremely encouraging and continuation of the exploration program is recommended.

Phase II should concentrate in the area outlined in Phase I and consist of linecutting, detail soil and rock sampling and geological mapping. Pending results, trenching may also be required.

Pending successful completion and encouraging results of Phase II, diamond drilling and/or overburden drilling will be recommended.

ITEMIZED COST ESTIMATE

PHYSICAL

Cat Trenching, blasting, explosives,
copco rock drill, etc.
(done in April 1984 - Kalum Lake Mining Group) \$12,017.00

SOIL SAMPLING

J. Holland - 12 days @ \$125/day 1,500.00
J. Dow - 20 days @ \$100/day 2,000.00
- 3 days @ \$150/day 450.00
(Mapping) - 10 days @ \$200/day 2,000.00
D. Howe (Supervision) - 2 days @ \$200/day 400.00
Report Writing and Drafting 2,000.00

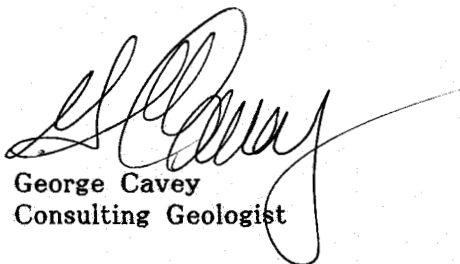
DISBURSEMENTS

Truck Rental 525.00
Gas 315.00
Freight 58.80
Kanata Travel 295.00
D. Howe - (Expense) 107.89
Neville Crosby 115.86
B.C. Telephone 50.74
Assays 8,701.43
Maps 20.00
\$18,540.62

QUALIFICATIONS

I, George Cavey, of 6891 Wiltshire Street, Vancouver, British Columbia hereby certify:

1. I am a graduate of the University of British Columbia (1976) and hold a BSc. degree in geology.
2. I am presently employed as a consulting geologist with OreQuest Consultants Ltd. of 404-595 Howe Street, Vancouver, British Columbia.
3. I have been employed in my profession by various mining companies for the past nine years.
4. I am a Fellow of the Geological Association of Canada.
5. I am a member of the Canadian Institute of Mining and Metallurgy.
6. The information contained in this report was obtained during the completion of the field work program supervised by OreQuest Consultants Ltd. in 1984.
7. Neither OreQuest Consultants Ltd. nor myself have direct or indirect interest in the property nor in the securities of Bradner Resources Ltd.
8. This report may be used by Bradner Resources Ltd. for all corporate purposes and including any public financing.



George Cavey
Consulting Geologist

DATED at Vancouver, British Columbia, this 19th day of November, 1984.

QUALIFICATIONS

I, Diane Howe, of 21394-126th Avenue, Maple Ridge, British Columbia hereby certify:

1. I am a graduate of the University of British Columbia (1980) and hold a BSc. degree in geology.
2. I am presently employed as a project geologist with OreQuest Consultants Ltd. of 404-595 Howe Street, Vancouver, British Columbia.
3. I have been employed in my profession by various mining companies for the past five years.
4. I am a member of the Canadian Institute of Mining.
5. The information contained in this report was obtained from data personally collected during the field program in April and September of 1984 and from the reports and files listed in the Bibliography.
6. Neither OreQuest Consultants Ltd. nor myself have direct or indirect interest in the property described nor in the securities of Bradner Resources Ltd.
7. This report may be used by Bradner Resources Ltd. for all corporate purposes and including any public financing.

Diane Howe
Project Geologist

DATED at Vancouver, British Columbia, this 19th day of November, 1984.

BIBLIOGRAPHY

B.C.D.M. ASSESSMENT REPORTS
#8299, #8393

B.C.D.M. MINISTRY OF MINES ANNUAL REPORT
1927-47, 1923-47, 1924-48, 1925-69, 1927-63, 1930-74

DUFFEL, S. and SOUTHER, J.G.
Geology of the Terrace Map Area, B.C. Memo 329; 1964, 117 pp.

G.S.C. PAPER
1956: 36-17-22

APPENDIX A

BRADJER

SAMPLE #	No ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	Hs ppm
05 0W	3	15	24	115	1.3	15	4
05 1W	5	31	27	140	.8	20	2
05 2W	3	24	40	119	1.2	50	10
05 3W	2	31	59	80	1.6	9400	20
05 4W	2	14	25	24	.4	5	2
05 5W	2	9	20	49	.3	15	2
05 6W	2	10	24	64	.8	5	2
05 7W	2	29	25	94	.6	10	10
05 8W	2	37	24	91	.4	5	4
05 9W	1	5	19	75	.4	nd	2
05 10W	1	10	10	35	.2	10	nd
05 11W	2	6	16	20	.4	20	2
05 12W	2	14	25	66	.7	15	4
05 13W	3	17	50	174	1.0	5	4
05 14W	2	15	33	80	1.2	10	nd
05 15W	3	18	24	113	1.6	15	10
05 16W	2	10	15	40	.9	40	10
05 17W	1	9	12	25	.6	50	nd
05 18W	2	4	16	41	.4	10	4
05 19W	1	2	9	9	.2	10	nd
05 20W	1	7	14	13	.7	5	2
1S 00W	2	24	20	220	.8	25	30
1S 01W	4	44	31	600	2.6	5	35
1S 02W	7	74	25	90	1.1	50	50
1S 03W	2	16	19	85	.7	20	2
1S 04W	7	25	20	143	.8	10	4
1S 05W	2	24	16	122	.8	15	2
1S 06W	2	12	15	49	.5	5	2
1S 07W	1	11	15	53	.7	20	4
1S 08W	2	25	20	70	1.0	nd	4
1S 09W	2	15	16	96	.2	5	10
1S 10W	2	14	19	55	.2	5	4
1S 320W	2	nd	16	15	.3	15	2
1S - 11W	5	95	23	166	nd	5	1.4
1S - 12W	3	21	27	144	20	5	.9
1S - 13W	3	20	26	105	10	10	1.0
1S - 14W	nd	9	15	19	2	10	.5
1S - 15W	2	15	15	26	4	5	.6
1S - 16W	4	19	23	70	15	10	.1
1S - 17W	3	10	14	74	2	15	.6
1S - 18W	2	11	15	27	4	5	.3
1S - 19W	1	12	13	54	4	5	.5

As

Ag

	No ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Pu ppb	Ag ppm
25 - 0W	2	22	20	102	40	nd	.4
25 - 1W	3	24	25	160	20	20	.6
25 - 2W	2	35	21	97	30	10	.8
25 - 3W	4	24	20	111	20	nd	.4
25 - 4W	3	20	25	226	10	nd	.4
25 - 5W	2	25	23	99	4	15	.2
25 - 6W	2	25	30	110	10	5	.3
25 - 7W	2	18	26	123	15	nd	.4
25 - 8W	2	13	18	100	10	5	.5
25 - 9W	3	27	18	150	10	nd	.3
25 - 10W	1	9	13	44	4	5	.2
25 - 11W	2	16	24	95	15	10	.3
25 - 12W	3	14	26	84	10	10	.6
25 - 13W	4	26	27	156	20	5	.5
25 - 14W	2	15	17	100	10	10	.7
25 - 15W	3	10	20	108	15	10	.4
25 - 16W	3	11	22	75	10	5	.5
25 - 17W	3	26	28	49	4	15	.1
25 - 18W	2	15	29	41	4	15	.2
25 - 19W	2	32	20	127	15	10	.5

0.20

L3S - 0W	3	15	21	165	35	5	.4
L3S - 1W	2	23	20	147	30	5	.3
L3S - 2W	1	7	9	35	2	10	.2
L3S - 3W	2	14	15	44	nd	10	.4
L3S - 4W	2	16	18	59	30	10	.6
L3S - 5W	1	9	10	27	10	10	.3
L3S - 6W	2	15	20	35	2	5	.7
L3S - 7W	5	72	31	95	50	10	1.6
L3S - 8W	17	18	15	75	20	nd	.3
L3S - 9W	3	14	24	100	10	10	.5
L3S - 10W	3	19	20	97	40	10	.6
L3S - 11W	2	22	17	73	30	10	.4
L3S - 12W	2	16	24	143	25	5	.1
L3S - 13W	2	17	21	87	20	5	.4
L3S - 14W	2	17	20	99	20	10	.4
L3S - 15W	2	10	23	166	15	10	.2
DETECTION LIMIT	1	1	2	1	2	5	

	Cu	Pb	Zn	As	Pu	Ag
L3S 16W	15	19	87	.5	nd	2
L3S 17W	9	20	75	.4	nd	2
L3S 18W	11	16	55	.4	10	2
L3S 19W	13	15	49	.4	nd	2
L3S 20W	11	13	36	.4	5	4

	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
L4S 0N	5	17	110	.2	nd	nd
L4S 1N	20	15	78	.4	15	4
L4S 2N	16	15	95	.5	nd	4
L4S 3N	40	41	205	.6	5	nd
L4S 4N	16	14	72	.6	nd	4
L4S 5N	17	15	98	.3	nd	2
L4S 6N	9	11	34	.2	nd	nd
L4S 7N	14	14	53	nd	nd	2
L4S 8N	9	12	68	.6	nd	4
L4S 9N	10	18	59	.6	nd	2
L4S 10N	15	16	71	.3	5	4
L4S 11N	11	27	116	0.1	5	15
L4S 12N	14	15	101	.5	nd	2
L4S 13N	12	11	49	.3	nd	2
L4S 14N	26	10	75	.6	nd	10
L4S 15N	12	11	64	.4	nd	4
L4S 16N	14	14	40	.4	nd	10
L4S 17N	15	16	45	.4	5	20
L4S 18N	10	12	50	.2	nd	4
L4S 19N	25	15	116	.4	nd	4
L4S 20N	7	12	30	.1	5	4
L5S 0N	12	22	114	.2	nd	2
L5S 1N	15	16	151	.5	nd	4
L5S 2N	35	18	136	.7	10	10
L5S 3N	20	20	120	.6	25	4
L5S 4N	24	15	77	.5	nd	4
L5S 5N	12	21	116	.7	5	2
L5S 6N	9	14	41	.2	5	2
L5S 7N	16	15	55	.4	10	10
L5S 8N	17	23	95	.3	5	15
L5S 9N	15	16	100	.5	nd	4
L5S 10N	21	21	144	.3	nd	4
L5S 11N	16	21	86	.3	10	20
L5S 12N	15	24	75	.6	10	2
L5S 13N	12	21	99	.6	nd	2
L5S 14N	15	17	86	.4	5	4
L5S 15N	20	15	86	.7	10	4
L5S 16N	15	21	39	.1	nd	4
L5S 17N	14	9	66	.3	nd	4
L5S 18N	11	21	71	.4	10	2
L5S 19N	12	15	60	.5	10	2
L5S 20N	22	17	123	.3	10	2

	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
L6S 0W	10	17	84	.3	nd	4
L6S 1W	10	20	90	.3	nd	4
L6S 2W	10	5	10	.6	10	2
L6S 3W	10	21	53	.7	10	4
L6S 4W	8	15	45	.3	5	4
L6S 5W	17	21	95	.3	10	4
L6S 6W	12	15	54	.5	5	2
L6S 7W	10	17	55	.5	5	2
L6S 8W	12	19	130	.3	nd	2
L6S 9W	11	12	78	.2	nd	2
L6S 10W	10	20	140	.6	15	4
L6S 11W	8	13	46	.4	5	2
L6S 12W	15	8	90	.8	nd	2
L6S 13W	36	4	130	.7	15	4
L6S 14W	21	28	123	.4	10	10
L6S 15W	10	25	53	.6	15	2
L6S 16W	9	15	60	.5	10	4
L6S 17W	14	20	40	.4	5	10
L6S 18W	14	19	55	.5	5	4
L6S 19W	10	20	46	.5	nd	2
L6S 20W	9	16	39	.3	nd	4
L7S 0W	5	4	18	.2	nd	2
L7S 1W	4	14	20	.1	5	2
L7S 3W	12	21	75	.4	15	4
L7S 3W	5	8	20	.2	10	2
L7S 4W	21	17	56	nd	5	4
L7S 5W	6	25	65	.8	5	4
L7S 6W	24	20	135	.5	5	10
L7S 7W	12	20	105	.7	10	4
L7S 8W	12	19	75	.5	5	4
L7S 9W	11	25	106	.2	5	4
L7S 10W	13	24	99	.6	5	4
L7S 11W	14	17	100	.2	nd	4
L7S 12W	10	19	42	.3	nd	15
L7S 13W	9	19	115	nd	nd	10
L7S 14W	11	24	60	.4	10	15
L7S 15W	14	17	105	.4	nd	4
L7S 16W	6	14	35	.2	5	2
L7S 17W	10	15	55	.1	10	2
L7S 18W	15	15	39	nd	10	4
L7S 19W	20	14	50	.5	50	4
L7S 20W	21	15	56	.2	nd	10

check
*2
no 2

SAMPLE #	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
BS 1W	5	5	7	8	1.0	10	2
BS 2W	3	27	16	57	1.3	5	15
BS 3W	3	14	24	124	1.1	5	4
BS 4W	2	14	20	65	.5	5	10
BS 5W	2	15	27	125	.6	10	4
BS 6W	2	5	29	48	.5	10	2
BS 7W	1	15	25	130	.3	10	4
BS 8W	3	9	17	61	.6	10	10
BS 9W	1	5	11	35	.2	5	2
LBS 12W		4	3	20	.1	10	nd
LBS 13W		12	35	95	.5	nd	4
LBS 14W		6	10	75	.2	nd	2
LBS 15W		15	22	120	.4	nd	15
LBS 16W		10	19	76	.7	5	2
LBS 17W		9	21	54	.7	nd	4
LBS 18W		19	22	44	.4	nd	2
LBS 19W		8	12	37	.1	nd	2
LBS 20W		16	15	48	.2	nd	2
9S 1W	3	19	16	100	.4	10	10
9S 2W	2	13	24	105	.6	10	4
9S 3W	2	21	25	125	.7	10	10
9S 4W	1	5	20	40	.3	20	2
9S 5W	2	12	21	110	.6	nd	4
9S 6W	2	9	20	84	.3	20	4
9S 7W	1	5	18	60	.3	15	2
9S 8W	2	10	15	50	.2	10	2
9S 9W	1	10	49	55	.7	15	4
L9S 12W		24	5	18	.6	nd	2
L9S 13W		4	14	28	.4	10	2
L9S 14W		5	15	44	.3	nd	4
L9S 15W		10	15	46	.2	5	15
L9S 16W		14	16	93	.2	5	4
L9S 17W		7	19	36	.4	10	4
L9S 18W		15	14	71	.3	nd	2
L9S 19W		6	5	35	.3	5	2
L9S 20W		9	13	66	nd	20	2

	No	Cu	Pb	Zn	Ag	Au	As
	ppm	ppm	ppm	ppm	ppm	ppb	ppm
105 1W	3	15	15	90	.5	5	4
105 2W	3	16	20	99	.9	10	4
105 3W	3	18	19	87	.5	10	4
105 4W	2	13	21	71	.7	5	4
105 5W	2	10	15	64	.2	nd	4
105 6W	3	11	21	115	.6	5	2
105 7W	3	9	20	75	.6	nd	4
105 12W	4	10	23	109	.8	5	4
105 13W	4	10	26	94	.6	10	10
105 14W	2	15	20	95	.6	nd	2
L105 16W		10	14	51	.6	10	4
L105 17W		13	23	102	.2	nd	10
L105 18W		6	19	40	.2	5	2
L105 1850W	18.5	21	10	54	.1	nd	10
L105 19W		6	7	27	.2	10	nd
L105 20W		10	10	34	.3	5	4
115 1W	2	6	21	65	.5	nd	2
115 2W	2	16	22	85	.4	nd	10
115 3W	2	10	16	85	.5	5	4
115 4W	3	8	21	100	.4	nd	2
115 5W	3	10	25	177	.5	nd	4
L115 13W		13	12	79	.3	5	4
L115 14W		9	9	34	.4	nd	10
L115 15W		16	15	71	.2	nd	4
L115 16W		15	13	51	.6	nd	4
L115 17W		15	15	71	.4	5	4
L115 18W		13	15	79	.6	nd	2
L115 1800W		21	14	60	.4	5	15
L115 19W		7	5	33	.3	nd	2
L115 20W		4	9	25	.2	nd	2
125 1W	2	16	19	69	.1	15	4
125 2W	2	9	24	75	.3	5	10
125 3W	2	9	10	61	.4	5	2
125 4W	3	15	14	75	.4	5	4
125 5W	2	7	19	94	.6	15	2
L125 13W		9	10	25	.3	5	4
L125 14W		23	19	150	.2	nd	15
L125 15W		10	12	99	.6	120	2
L125 16W		15	19	81	.1	5	4
L125 17W		5	11	20	.4	nd	nd
L125 18W		4	6	15	.5	nd	2
L125 19W		9	19	35	.3	5	2
L125 20W		25	12	50	.1	5	10

check
no #15

check
18A,B

	Mo (ppm)	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
13S 1W	1	2	5	18	.4	5	2
13S 2W	1	9	15	62	.5	nd	18
13S 3W	5	14	19	142	.4	nd	4
13S 4W	2	14	16	67	.5	5	4
13S 13W	3	15	22	58	.6	5	15
13S 14W	3	5	19	46	.8	5	2
13S 15W	2	11	48	49	1.8	5	4
13S 16W	5	18	26	37	.6	5	2
13S 17W	2	8	21	94	.6	nd	2
14S 1W	3	26	25	127	1.8	5	18
14S 2W	2	18	15	47	.4	5	2
14S 3W	3	25	24	98	.6	5	18
14S 4W	2	9	21	65	.4	5	2
14S 18W	3	12	24	45	.4	18	18
14S 11W	3	15	14	34	.3	5	4
14S 12W	3	8	19	58	.4	nd	2
14S 13W	3	13	21	55	.1	18	28
14S 14W	3	6	21	46	1.4	18	2
14S 15W	3	6	16	53	.4	nd	2
14S 16W	3	18	25	198	.7	18	4
14S 17W	3	18	28	88	.7	18	2
15S 1W	2	5	28	65	.3	5	4
15S 2W	2	15	21	183	.3	nd	4
15S 3W	3	6	18	67	.2	nd	2
15S 18W	2	9	19	38	.5	18	4
15S 11W	2	4	18	16	.6	5	2
15S 12W	2	5	19	88	.4	5	2
15S 13W	4	7	18	25	.4	5	4
15S 14W	2	7	25	38	.4	5	4

	Mo	Cu	Pb	Zn	As	Ag	Au
16S 0W	3	14	18	79	.6	nd	4
16S 1W	3	11	22	75	.6	10	2
16S 2W	7	10	18	100	.4	5	4
16S 3W	7	9	23	114	.5	nd	4
16S 0W	3	14	25	44	.5	nd	4
16S 9W	4	12	30	51	.7	15	2
16S 10W	4	16	18	64	.3	10	4
16S 11W	2	10	16	49	.8	10	4
16S 12W	4	6	21	104	.4	5	10
16S 13W	4	10	23	69	.6	5	4
16S 14W	3	10	25	88	.4	10	10
17S 0W	1	6	10	69	.5	10	2
17S 1W	5	14	20	103	.5	10	10
17S 2W	3	10	25	134	.6	10	4
17S 3W	4	16	35	145	.4	5	4
17S 7W	2	7	17	30	.4	5	2
17S 8W	3	11	22	50	.4	5	2
17S 9W	3	10	15	31	.6	5	10
17S 10W	3	14	15	33	.3	5	20
17S 11W	2	5	15	25	.7	15	4
18S 0W	2	13	13	137	.7	5	2
18S 1W	1	24	14	75	.2	nd	4
18S 2W	2	10	20	149	.6	5	2
18S 3W	2	7	14	108	.3	nd	4
18S 8W	3	14	21	53	.5	10	10
18S 9W	3	10	19	49	.4	5	4
18S 10W	2	11	14	46	.3	5	4
18S 11W	4	14	21	70	.3	nd	4

Check
FD#4

	No ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
1N 0W	(20)	21	30	124	.4	10	2
1N 1W	8	34	29	80	.5	(10)	4
1N 2W	(19)	14	31	56	.4	nd	2
1N 3W	3	15	30	78	.7	10	4
1N 5W	2	35	19	35	.2	5	nd
1N 6W	2	10	31	149	.4	10	2
1N 7W	2	14	27	39	.6	(5)	4
1N 8W	3	15	25	60	.4	10	4
1N 9W	2	25	30	95	.7	5	10
1N 10W	2	14	36	39	.5	10	4
1N 11W	2	28	30	88	.4	15	15
1N 12W	2	10	30	75	.8	15	10
1N 13W	4	7	21	37	.4	10	2
1N 14W	3	39	40	144	(3.1)	20	10
1N 15W	2	20	32	80	.6	10	10
1N 16W	3	10	39	99	.6	5	15
1N 17W	2	14	24	35	.6	5	4
1N 18W	1	4	16	11	.4	5	nd
1N 19W	1	4	7	5	.1	nd	nd
1N 20W	1	5	7	10	.8	10	2
2N 0W	(10)	9	14	58	.3	nd	nd
2N 1W	9	25	20	135	.8	nd	4
2N 2W	2	12	25	140	.7	5	15
2N 3W	1	15	17	75	(1.0)	10	2
2N 4W	2	5	2	9	.5	nd	4
2N 5W	1	6	16	69	.3	5	2
2N 6W	2	17	20	75	.4	(150)	15
2N 7W	2	20	25	85	.2	5	4
2N 8W	1	5	10	35	.3	nd	nd
2N 9W	1	10	5	25	.7	5	4
2N 10W	2	26	25	(229)	(2.0)	nd	10
2N 11W	2	15	24	177	.5	nd	4
2N 12W	3	15	18	55	1.2	nd	4
2N 13W	2	31	16	119	nd	nd	(80)
2N 14W	2	15	10	25	.2	nd	2
2N 15W	2	9	22	29	.9	nd	4
2N 16W	2	21	15	34	.3	nd	10
2N 17W	2	10	15	30	.1	15	4
2N 18W	3	15	19	41	.7	5	2
2N 19W	2	10	16	9	.6	10	nd
2N 20W	2	25	22	56	.5	10	10

	No ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
3N 00W	1	15	15	75	.5	15	4
3N 01W	1	20	15	130	.1	10	15
3N 02W	2	10	15	50	.3	5	2
3N 03W	2	25	19	75	.4	nd	15
3N 04W	2	5	11	45	1.0	10	4
3N 05W	nd	3	5	15	.1	15	2
3N 06W	2	27	15	90	nd	20	10
3N 07W	1	13	13	60	.5	20	4
3N 08W	3	16	45	130	.7	35	10
3N 09W	1	10	9	24	.7	10	nd
3N 10W	2	11	10	23	.2	10	nd
3N 11W	2	8	9	29	.2	20	4
3N 12W	3	14	14	50	.5	10	10
3N 13W	3	14	14	85	.3	20	15
3N 14W	nd	5	5	17	.2	nd	2
3N 15W	1	2	2	5	.2	10	2
3N 16W	nd	10	11	39	nd	10	10
3N 17W	nd	8	6	13	nd	5	10
3N 18W	nd	3	1	9	nd	5	2
3N 19W	2	10	13	35	.7	5	4
3N 20W	2	8	18	54	.3	nd	15
4N 00W	2	8	14	64	.2	nd	10
4N 01W	1	10	14	61	.3	nd	4
4N 02W	1	6	6	21	.4	5	2
4N 03W	2	7	8	59	.3	10	2
4N 04W	1	10	5	52	.3	nd	nd
4N 05W	1	5	10	14	.4	10	2
4N 06W	1	14	15	80	.6	25	10
4N 07W	nd	2	15	20	.4	10	2
4N 08W	2	24	25	220	1.1	10	4
4N 09W	1	30	40	38	1.3	10	nd
4N 10W	2	10	16	77	1.0	10	4
4N 11W	1	5	6	20	.4	5	nd
4N 12W	1	4	3	12	.2	15	nd
4N 13W	2	11	11	39	.2	10	2
4N 14W	2	14	23	50	.7	5	15
4N 15W	2	13	10	35	1.5	nd	2
4N 16W	4	20	18	45	1.3	5	4
4N 17W	7	12	30	38	.3	25	15
4N 18W	1	6	9	15	.2	nd	2
4N 19W	1	nd	5	5	.4	10	2
4N 20W	2	14	13	28	.2	nd	4

	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
5N 00W	1	5	10	110	.5	50	20
5N 01W	2	8	11	1	.8	20	4
5N 02W	1	10	14	84	.5	10	4
5N 03W	1	23	24	114	1.4	5	10
5N 04W	1	13	24	100	1.2	nd	10
5N 05W	2	5	9	25	.3	nd	2
5N 07W	1	15	16	90	.5	nd	4
5N 08W	1	4	5	10	.2	50	nd
5N 09W	nd	14	14	55	1.2	15	15
5N 10W	1	35	96	200	1.1	5	2
5N 11W	4	25	25	90	.7	5	50
5N 12W	1	3	nd	5	nd	5	2
5N 13W	1	6	10	35	.1	nd	2
5N 14W	2	15	15	62	2.1	nd	4
5N 15W	4	15	33	96	1.0	nd	30
5N 16W	3	15	24	42	.6	nd	4
5N 17W	2	8	9	25	.3	nd	2
5N 18W	7	14	20	53	.6	10	2
5N 19W	3	19	15	55	.3	nd	4
5N 20W	2	8	11	25	.3	20	4
6N 00W	1	5	9	33	.3	5	nd
6N 01W	2	19	19	95	.7	nd	2
6N 02W	2	10	20	80	.5	nd	4
6N 03AW	3	31	20	144	.2	50	10
6N 03BW	3	13	20	184	.3	nd	4
6N 04W	1	10	14	55	.6	320	10
6N 05W	4	30	160	280	1.5	100	50
6N 06W	1	11	14	70	.6	10	10
6N 07W	1	5	10	30	.3	nd	2
6N 08W	1	15	15	59	.5	5	4
6N 09W	2	14	20	133	.2	15	2
6N 10W	2	15	15	50	.4	nd	4
6N 11W	3	23	15	70	.6	nd	2
6N 12W	1	16	15	75	.5	nd	2
6N 13AW	2	10	15	50	.4	nd	10
6N 13BW	3	30	17	159	.9	nd	2
6N 15W	3	10	19	80	.6	nd	2
6N 16W	2	15	15	55	.4	nd	4
6N 17W	6	24	19	66	.5	5	4
6N 18W	2	10	14	24	.3	15	2
6N 19W	3	15	15	36	.6	10	10
6N 20W	4	17	16	40	.3	nd	15

check

check

SAMPLE #

Cu ppm Pb ppm Zn ppm Ag ppm Au ppb As ppm

L7N 0W 5 5 28 .3 5 nd
 L7N 1W 8 16 114 .3 nd 4
 L7N 2W 14 24 93 .4 nd 2
 L7N 3W 18 24 59 .4 nd 4
 L7N 4W 14 19 39 .6 nd 2

L7N 5W 15 15 91 .8 10 4
 L7N 6W 5 10 35 .5 5 nd
 L7N 7W 4 11 20 .2 nd 2
 L7N 8W 25 15 63 .3 45 10
 L7N 9W 5 10 20 .2 nd nd

L7N 10W 25 44 190 .7 10 2
 L7N 11W 4 5 20 .2 10 2
 L7N 12W 4 20 12 .2 5 nd
 L7N 13W 13 10 46 .6 20 2
 L7N 14W Mo 14 8 41 .6 5 2

7N 15W 2 15 15 39 .4 5 2
 7N 16W nd 7 7 16 .3 10 2
 7N 17W 3 30 20 60 1.0 nd 4

7N 18W 5 13 12 42 .3 5 10
 7N 19W 5 10 10 41 .3 nd 10
 7N 20W 4 10 9 38 .3 10 4

L8N 0W 12 12 55 .4 nd 4
 L8N 1W 20 18 45 .5 nd 2
 L8N 2W 20 19 120 .5 10 10
 L8N 3W 6 14 66 .3 nd nd
 L8N 4W 21 35 91 .5 nd 10

L8N 5W 10 15 70 .5 10 4
 L8N 6W 8 9 40 .5 nd 2
 L8N 7W 4 2 15 .4 nd nd
 L8N 8W 3 6 15 .4 nd 2
 L8N 9W 12 15 93 1.1 nd 4

L8N 10W 9 24 25 .7 5 2
 L8N 11W 10 10 31 .4 nd 2
 L8N 12W 15 25 69 .5 nd 10
 L8N 13W 11 19 34 1.0 nd 2
 L8N 14W 10 9 24 .6 nd nd

L8N 15W 7 5 11 .8 5 2
 L8N 16W 10 15 35 .1 5 4
 L8N 17W 10 15 33 .2 nd 4
 L8N 18W 9 13 24 .1 5 2
 L8N 19W 21 20 73 .4 nd 15

L8N 20W 17 15 42 .8 35 2

	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
L9N 0W	10	10	61	.7	5	2
L9N 1W	11	7	48	.5	nd	2
L9N 2W	12	11	65	.4	nd	2
L9N 5W	11	10	40	.2	30	10
L9N 6W	9	12	25	nd	10	2
L9N 7W	25	16	70	.4	nd	10
L9N 8W	24	7	8	.5	nd	nd
L9N 9W	13	12	75	.6	25	2
L9N 10W	20	20	71	1.0	nd	4
L9N 11W	14	11	44	.4	5	4
L9N 12W	19	14	97	.4	nd	4
L9N 13W	15	17	50	.2	nd	2
L9N 14W	10	10	24	.1	nd	10
L9N 15W	4	6	7	.3	nd	nd
L9N 16W	3	5	4	.2	nd	nd
L9N 17W	18	10	33	.3	nd	2
L9N 18W	27	12	55	nd	5	4
L9N 19W	49	26	71	1.3	nd	4
L9N 20W	6	2	10	.2	nd	nd

	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm
10N 1W	2	9	20	39	1.0	15	4
10N 2W	2	10	15	90	.1	5	4
10N 3W	2	8	10	44	1.0	nd	4
10N 4W	5	25	22	65	2.1	nd	2
10N 5W	3	16	15	52	.4	5	4
10N 6W	3	12	15	74	.7	10	2
10N 7W	1	9	14	45	.3	10	4
10N 8W	2	16	15	57	.3	10	4
10N 9W	3	7	17	52	.3	10	2
10N 10W	1	9	14	45	.4	20	2
10N 11W	2	15	21	41	.4	5	4
10N 12W	3	24	16	68	.3	20	20
10N 13W	2	10	14	85	.6	10	2
10N 14W	4	11	19	71	.7	10	4
10N 15W	3	10	16	53	.8	5	4
10N 16W	4	10	21	89	.4	20	2
10N 17W	3	13	15	89	.7	15	4
10N 18W	3	10	15	50	.4	5	4
10N 19W	4	10	16	71	.2	20	4
10N 20W	4	10	19	29	.7	5	2

	Ni ppm	Cu ppm	Pb ppm	Zn ppm	Hg ppm	Mn ppb	MS ppm
12N 2W	2	5	15	43	.4	25	2
12N 3W	2	13	14	50	.6	5	4
12N 4W	3	9	20	45	1.2	nd	2
12N 5W	3	11	20	49	.4	5	4
12N 6W	2	14	15	65	.2	5	4
12N 7W	4	8	19	56	.2	5	nd
12N 8W	1	6	9	34	.3	5	2
12N 9W	5	37	31	134	.7	5	10
12N 10W	2	6	15	78	.6	5	2
12N 11W	3	10	18	50	.4	10	4
12N 11.50W	3	14	16	100	1.1	nd	4
12N 12W	3	9	20	45	.2	nd	10
12N 13W	4	13	20	36	.4	nd	2
12N 14W	3	10	18	47	.2	5	4
12N 15W	3	10	17	63	.5	5	4
12N 16W	6	16	15	39	.8	5	10
12N 17W	5	36	19	45	.6	5	4
12N 18W	2	11	12	50	.1	5	4
12N 19W	6	17	12	45	.5	10	4
12N 20W	7	14	15	50	.6	10	10
14N 13W	3	5	6	15	.3	nd	nd
14N 13.50W	3	5	7	28	.4	10	2
14N 14W	2	2	16	14	.2	nd	2
14N 14.50W	3	9	18	25	.7	5	4
14N 15W	3	9	16	53	.8	5	2
14N 15.50W	3	9	20	35	.8	5	4
14N 16W	3	10	15	41	.6	5	2
14N 16.50W	3	8	15	30	1.4	5	2
14N 17W	3	6	14	17	.5	10	2
14N 17.50W	4	26	21	74	.9	nd	10
14N 18W	4	20	16	41	.8	10	4
14N 18.50W	2	15	14	27	.4	nd	4
14N 19W	3	9	14	60	.3	nd	4
14N 19.50W	4	14	17	53	.5	10	4
14N 20W	4	15	15	55	.4	10	4
16N 13.50W	2	5	15	24	.4	5	nd
16N 14W	1	10	11	22	.4	nd	2
16N 14.50W	2	6	15	24	.2	10	4
16N 15W	1	11	22	47	.8	10	10
16N 15.50W	2	10	23	54	.7	10	4
16N 15.60W	4	14	20	50	.2	5	10
16N 16W	2	5	17	35	.3	10	4
16N 16.50W	3	7	20	42	.4	nd	4
16N 17W	3	6	15	27	.5	10	2
16N 17.50W	3	15	13	45	.7	10	2
16N 18W	4	8	25	25	.5	5	10
16N 18.50W	3	15	15	55	.4	5	4
16N 19W	4	10	21	51	.7	10	4
16N 19.30W	2	14	10	50	.3	5	50
16N 19.50W	3	15	16	92	.5	10	nd

	Mo	Cu	Pb	Zn	Ag	Au	As
18N 13W	2	8	6	17	.3	10	2
18N 13.50W	3	15	15	35	.4	15	4
18N 14W	3	14	16	40	.3	10	4
18N 14.50W	3	16	12	33	.5	5	2
18N 15W	2	54	16	79	.2	5	15
18N 15.50W	1	9	17	40	.3	5	2
18N 16W	2	8	12	19	.2	20	30
18N 16.50W	3	10	14	25	.5	5	50
18N 17W	1	11	14	25	.4	nd	60
18N 17.50W	2	14	11	37	.4	nd	4
18N 18W	4	12	14	46	.6	5	2
18N 18.50W	1	7	9	25	.1	10	30
18N 19W	2	10	11	79	.1	10	25
18N 19.50W	4	25	26	64	.3	10	40
18N 20W	4	36	22	80	.3	nd	30
20N 13W	3	12	6	14	.5	5	2

ROCKS

Sample LOCATION

SAMPLE #	Mo ppm	Cu ppm	Pb ppm	Zn ppm	As ppm	Au ppb	Ag ppm	Sample LOCATION
KALUM 1	1	24	18	78	18	5	.5	70m N of 200S, 2000W
KALUM 2	2	45	17	123	60	nd	.6	1100S, 1100W
KALUM 3	1	11	12	61	20	5	.4	1100S, 1180W
DETECTION LIMIT	1	1	2	1	2	5	0.1	

SAMPLE #	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm	Sample LOCATION
C - 1	600	65	14	8.8	640	50	QZ - SOUTH SHOULDR.
C - 2	5800	1500	115	80.1	16700	100	#1 UCW - DUMP (BAV)
C - 3	69	29	39	4.2	70	4	1400S, 2100W (AQZ)
C - 4	10	5	71	.5	5	nd	4100W, 4130W (AEG) QZ
C - 5	6	nd	35	.7	nd	4	4100S, 8150W (") QZ
C - 6	39	26	124	.6	15	2	0100, 2100W (")
C - 7	65	15	32	.8	5	nd	2100N, 10150W (RHY?)
C - 8	12	13	71	.4	5	nd	0100, 9100W (AEG) QZ
C - 9	22	9	74	.5	nd	10	1100W, 15130W (")
C - 10	18	4	53	.2	5	10	2100N, 13100W (") QZ
C - 11	13	10	67	.4	nd	10	5100N, 15150W (") QZ
C - 13	127	430	19	89.2	1575	2	BAV #2
C - 14	1520	850	2	199.2	6580	15	"

	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Au ppb	As ppm	Sample LOCATION
OC 1N 1450W (ROCK)	1	35	13	84	.2	nd	10	1100W, 14150W

File B:BRADNER.VGC Threshold/Anomaly Statistics
GOLD

B:BRADNER.VGC / col 7	Normal	Log-xform	Un-xform
No. Samples	387		
Min	5.00		
Max	100.00		
Mean	10.84	0.93	8.59
Standard Deviation	10.73	0.26	1.83
Threshold	32.29	1.46	28.64
Anomalous	43.02	1.72	52.29
Very Anomalous	53.74	1.98	95.47
Upper Discard Limit	100.00		

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)		
(1)	45.00	** ANOM	1.65	*	thresh
(3)	50.00	** ANOM	1.70	*	thresh
(15)	40.00	* thresh	1.60	*	thresh
(16)	50.00	** ANOM	1.70	*	thresh
(21)	40.00	* thresh	1.60	*	thresh
(25)	35.00	* thresh	1.54	*	thresh
(40)	60.00	*** V ANOM	1.78	**	ANOM
(65)	35.00	* thresh	1.54	*	thresh
(68)	80.00	*** V ANOM	1.90	**	ANOM
(90)	50.00	** ANOM	1.70	*	thresh
(94)	60.00	*** V ANOM	1.78	**	ANOM
(102)	80.00	*** V ANOM	1.90	**	ANOM
(103)	100.00	*** V ANOM	2.00	***	V ANOM
(261)	45.00	** ANOM	1.65	*	thresh
(273)	35.00	* thresh	1.54	*	thresh
(275)	30.00		1.48	*	thresh
(329)	50.00	** ANOM	1.70	*	thresh
(377)	30.00		1.48	*	thresh

Note: 10 value(s) not included (exceed limit of 100)

File B:BRADNER.VGC Threshold/Anomaly Statistics
SILVER

B:BRADNER.VGC / col 6 Normal Log-xform Un-xform

	Normal	Log-xform	Un-xform
No. Samples	563		
Min	0.10		
Max	3.10		
Mean	0.51	-0.37	0.43
Standard Deviation	0.33	0.26	1.82
Threshold	1.16	0.15	1.41
Anomalous	1.49	0.41	2.56
Very Anomalous	1.82	0.67	4.65

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)	
(1)	1.30	* thresh	0.11	
(3)	1.20	* thresh	0.08	
(4)	1.60	** ANOM	0.20	* thresh
(15)	1.20	* thresh	0.08	
(16)	1.60	** ANOM	0.20	* thresh
(35)	3.10	*** V ANOM	0.49	** ANOM
(43)	2.60	*** V ANOM	0.41	** ANOM
(64)	2.00	*** V ANOM	0.30	* thresh
(66)	1.20	* thresh	0.08	
(100)	1.30	* thresh	0.11	
(106)	1.50	** ANOM	0.18	* thresh
(107)	1.30	* thresh	0.11	
(115)	1.40	* thresh	0.15	
(116)	1.20	* thresh	0.08	
(120)	1.20	* thresh	0.08	
(124)	2.10	*** V ANOM	0.32	* thresh
(137)	1.60	** ANOM	0.20	* thresh
(161)	1.30	* thresh	0.11	
(181)	2.10	*** V ANOM	0.32	* thresh
(215)	1.20	* thresh	0.08	
(254)	1.40	* thresh	0.15	
(270)	1.40	* thresh	0.15	
(393)	1.30	* thresh	0.11	
(520)	1.40	* thresh	0.15	
(556)	1.60	** ANOM	0.20	* thresh

Class % vs. Log10(PPM): B:BRADNER.VGC

SILVER

upper			F R E Q U E N C Y (%)				
Log10	Class%	Cum%	1	2	3	4	5
0.03	94.32	94.32	*****>>				
0.13	3.20	97.51	T***
0.23	1.60	99.11	**
0.33	0.53	99.64	A*
0.43	0.18	99.82
0.53	0.18	100.00

Cumulative Percent vs. Log(PPM): B:BRADNER.VGC

SILVER

upper			P R O B A B I L I T Y (C U M %)												
PPM	Log10	Cum%	.01	.1	.5	2	10	30	50%	70	90	98	99	999	9999
1.1	0.03	94.32	:	:	:	:	:	:		:	:	+	:	:	:
1.3	0.13	97.51	:	:	:	:	:	:		:	:	+	:	:	:
1.7	0.23	99.11	:	:	:	:	:	:		:	:	:	+	:	:
2.1	0.33	99.64	:	:	:	:	:	:		:	:	:	:	+	:
2.7	0.43	99.82	:	:	:	:	:	:		:	:	:	:	:	+
3.4	0.53	100.00	:	:	:	:	:	:		:	:	:	:	:	+
			.01	.1	.5	2	10	30	50%	70	90	98	99	999	9999

File B:BRADNER.VGC Threshold/Anomaly Statistics
ARSENIC

B:BRADNER.VGC / col 8 Normal Log-xform Un-xform

	Normal	Log-xform	Un-xform
No. Samples	536		
Min	2.00		
Max	80.00		
Mean	6.88	0.66	4.52
Standard Deviation	8.71	0.36	2.27
Threshold	24.29	1.37	23.40
Anomalous	33.00	1.73	53.21
Very Anomalous	41.71	2.08	120.99

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)	
(35)	30.00	* thresh	1.48	* thresh
(36)	35.00	** ANOM	1.54	* thresh
(37)	50.00	*** V ANOM	1.70	* thresh
(58)	80.00	*** V ANOM	1.90	** ANOM
(110)	40.00	** ANOM	1.60	* thresh
(114)	30.00	* thresh	1.48	* thresh
(125)	50.00	*** V ANOM	1.70	* thresh
(280)	50.00	*** V ANOM	1.70	* thresh
(308)	30.00	* thresh	1.48	* thresh
(309)	50.00	*** V ANOM	1.70	* thresh
(310)	60.00	*** V ANOM	1.78	** ANOM
(313)	30.00	* thresh	1.48	* thresh
(314)	35.00	** ANOM	1.54	* thresh
(315)	40.00	** ANOM	1.60	* thresh
(316)	30.00	* thresh	1.48	* thresh
(502)	40.00	** ANOM	1.60	* thresh
(504)	30.00	* thresh	1.48	* thresh
(522)	35.00	** ANOM	1.54	* thresh
(523)	30.00	* thresh	1.48	* thresh
(525)	30.00	* thresh	1.48	* thresh
(528)	50.00	*** V ANOM	1.70	* thresh
(531)	40.00	** ANOM	1.60	* thresh
(532)	30.00	* thresh	1.48	* thresh
(533)	25.00	* thresh	1.40	* thresh

ARSENIC

upper			F R E Q U E N C Y (%)											
Log10	Class%	Cum%	+	1	2	3	4	5	6	7	8	9	+	
0.03	0.00	0.00	:	.	:	.	:	.	:	.	:	.	:	
0.13	0.00	0.00	:	.	:	.	:	.	:	.	:	.	:	
0.23	0.00	0.00	:	.	:	.	:	.	:	.	:	.	:	
0.33	33.96	33.96	:	*****										:
0.43	0.00	33.96	:	.	:	.	:	.	:	.	:	.	:	
0.53	0.00	33.96	:	.	:	.	:	.	:	.	:	.	:	
0.63	37.87	71.83	:	M*****										:
0.73	0.00	71.83	:	.	:	.	:	.	:	.	:	.	:	
0.83	0.00	71.83	:	.	:	.	:	.	:	.	:	.	:	
0.93	0.00	71.83	:	.	:	.	:	.	:	.	:	.	:	
1.03	15.49	87.31	:	*****										:
1.13	0.00	87.31	:	.	:	.	:	.	:	.	:	.	:	
1.23	5.60	92.91	:	*****										:
1.33	2.61	95.52	:	T***										:
1.43	0.19	95.71	:	.	:	.	:	.	:	.	:	.	:	
1.53	1.68	97.39	:	**										:
1.63	1.31	98.69	:	A*										:
1.73	0.93	99.63	:	*										:
1.83	0.19	99.81	:	.	:	.	:	.	:	.	:	.	:	
1.93	0.19	100.00	:	.	:	.	:	.	:	.	:	.	:	

Cumulative Percent vs. Log(PPM): B:BRADNER.VGC

ARSENIC

upper			P R O B A B I L I T Y (C U M %)												
PPM	Log10	Cum%	.01	.1	.5	2	10	30	50%	70	90	98	99	999	9999
1.7	0.23	0.00	:	:	:	:	:	:	:	:	:	:	:	:	:
2.1	0.33	33.96	:	:	:	:	:	+	:	:	:	:	:	:	:
2.7	0.43	33.96	:	:	:	:	:	:	:	:	:	:	:	:	:
3.4	0.53	33.96	:	:	:	:	:	:	:	:	:	:	:	:	:
4.3	0.63	71.83	:	:	:	:	:	:	:	+	:	:	:	:	:
5.4	0.73	71.83	:	:	:	:	:	:	:	:	:	:	:	:	:
6.8	0.83	71.83	:	:	:	:	:	:	:	:	:	:	:	:	:
8.5	0.93	71.83	:	:	:	:	:	:	:	:	:	:	:	:	:
11	1.03	87.31	:	:	:	:	:	:	:	:	+	:	:	:	:
13	1.13	87.31	:	:	:	:	:	:	:	:	:	:	:	:	:
17	1.23	92.91	:	:	:	:	:	:	:	:	:	+	:	:	:
21	1.33	95.52	:	:	:	:	:	:	:	:	:	:	+	:	:
27	1.43	95.71	:	:	:	:	:	:	:	:	:	:	+	:	:
34	1.53	97.39	:	:	:	:	:	:	:	:	:	:	+	:	:
43	1.63	98.69	:	:	:	:	:	:	:	:	:	:	+	:	:
54	1.73	99.63	:	:	:	:	:	:	:	:	:	:	+	:	:
68	1.83	99.81	:	:	:	:	:	:	:	:	:	:	+	:	:
85	1.93	100.00	:	:	:	:	:	:	:	:	:	:	+	:	+

File B:BRADNER.VGC Threshold/Anomaly Statistics
ZINC

B:BRADNER.VGC / col 5	Normal	Log-xform	Un-xform
No. Samples	576		
Min	1.00		
Max	286.00		
Mean	66.81	1.74	54.33
Standard Deviation	41.04	0.31	2.02
Threshold	148.89	2.35	222.19
Anomalous	189.93	2.65	449.33
Very Anomalous	230.97	2.96	908.69
Upper Discard Limit	500.00		

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)	
(14)	174.00	* thresh	2.24	
(27)	149.00	* thresh	2.17	
(42)	220.00	** ANOM	2.34	
(63)	229.00	** ANOM	2.36	* thresh
(64)	177.00	* thresh	2.25	
(103)	220.00	** ANOM	2.34	
(125)	200.00	** ANOM	2.30	
(140)	184.00	* thresh	2.26	
(142)	286.00	*** V ANOM	2.46	* thresh
(151)	159.00	* thresh	2.20	
(217)	177.00	* thresh	2.25	
(277)	198.00	** ANOM	2.30	
(340)	149.00	* thresh	2.17	
(357)	190.00	** ANOM	2.28	
(410)	205.00	** ANOM	2.31	
(429)	151.00	* thresh	2.18	
(525)	150.00	* thresh	2.18	
(532)	166.00	* thresh	2.22	
(542)	160.00	* thresh	2.20	
(545)	226.00	** ANOM	2.35	* thresh
(550)	150.00	* thresh	2.18	
(554)	156.00	* thresh	2.19	
(561)	165.00	* thresh	2.22	
(576)	166.00	* thresh	2.22	

Note: 2 value(s) not included (exceed limit of 500)

Class % vs. Log10(PPM): B:BRADNER.VGC

ZINC

upper			F R E Q U E N C Y (%)				
Log10	Class%	Cum%	1	2	3	4	5
0.03	0.17	0.17	:	:	:	:	:
0.13	0.00	0.17	:	:	:	:	:
0.23	0.00	0.17	:	:	:	:	:
0.33	0.00	0.17	:	:	:	:	:
0.43	0.00	0.17	:	:	:	:	:
0.53	0.00	0.17	:	:	:	:	:
0.63	0.17	0.35	:	:	:	:	:
0.73	0.69	1.04	*	:	:	:	:
0.83	0.00	1.04	:	:	:	:	:
0.93	0.52	1.56	*	:	:	:	:
1.03	1.39	2.95	*	:	:	:	:
1.13	1.04	3.99	*	:	:	:	:
1.23	2.08	6.08	**	:	:	:	:
1.33	3.30	9.38	***	:	:	:	:
1.43	4.86	14.24	*****	:	:	:	:
1.53	4.69	18.92	*****	:	:	:	:
1.63	11.98	30.90	*****	:	:	:	:
1.73	13.72	44.62	M*****	:	:	:	:
1.83	13.89	58.51	*****	:	:	:	:
1.93	14.41	72.92	*****	:	:	:	:
2.03	12.67	85.59	*****	:	:	:	:
2.13	7.47	93.06	*****	:	:	:	:
2.23	4.69	97.74	*****	:	:	:	:
2.33	1.39	99.13	T*	:	:	:	:
2.43	0.69	99.83	*	:	:	:	:
2.53	0.17	100.00	:	:	:	:	:

File B:BRADNER.VGC Threshold/Anomaly Statistics
LEAD

B:BRADNER.VGC / col 4 Normal Log-xform Un-xform

	Normal	Log-xform	Un-xform
No. Samples	576		
Min	1.00		
Max	160.00		
Mean	17.78	1.20	15.91
Standard Deviation	9.90	0.21	1.64
Threshold	37.59	1.63	42.64
Anomalous	47.49	1.84	69.82
Very Anomalous	57.39	2.06	114.30

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)		
(3)	40.00	* thresh	1.60		
(4)	59.00	*** V ANOM	1.77	* thresh	
(14)	50.00	** ANOM	1.70	* thresh	
(35)	40.00	* thresh	1.60		
(37)	39.00	* thresh	1.59		
(83)	45.00	* thresh	1.65	* thresh	
(105)	40.00	* thresh	1.60		
(126)	96.00	*** V ANOM	1.98	** ANOM	
(142)	160.00	*** V ANOM	2.20	*** V ANOM	
(182)	49.00	** ANOM	1.69	* thresh	
(249)	40.00	* thresh	1.60		
(357)	44.00	* thresh	1.64	* thresh	
(410)	41.00	* thresh	1.61		

Class % vs. Log10(PPM): B:BRADNER.VGC

LEAD

upper			F R E Q U E N C Y (%)				
Log10	Class%	Cum%	1	2	3	4	5
0.03	0.17	0.17	:	:	:	:	:
0.13	0.00	0.17	:	:	:	:	:
0.23	0.00	0.17	:	:	:	:	:
0.33	0.69	0.87	:*	:	:	:	:
0.43	0.00	0.87	:	:	:	:	:
0.53	0.35	1.22	:	:	:	:	:
0.63	0.35	1.56	:	:	:	:	:
0.73	2.60	4.17	:***	:	:	:	:
0.83	1.56	5.73	**	:	:	:	:
0.93	2.08	7.81	**	:	:	:	:
1.03	7.64	15.45	:*****	:	:	:	:
1.13	8.16	23.61	M*****	:	:	:	:
1.23	27.78	51.39	:*****	:	:	:	:
1.33	27.08	78.47	:*****	:	:	:	:
1.43	13.89	92.36	:*****	:	:	:	:
1.53	4.69	97.05	T*****	:	:	:	:
1.63	1.74	98.78	**	:	:	:	:
1.73	0.69	99.48	*	:	:	:	:
1.83	0.17	99.65	A	:	:	:	:
1.93	0.00	99.65	:	:	:	:	:
2.03	0.17	99.83	V	:	:	:	:
2.13	0.00	99.83	:	:	:	:	:
2.23	0.17	100.00	:	:	:	:	:

Cumulative Percent vs. Log(PPM): B:BRADNER.VGC

LEAD

upper			P R O B A B I L I T Y (C U M %)												
PPM	Log10	Cum%	.01	.1	.5	2	10	30	50%	70	90	98	99	999	9999
1.1	0.03	0.17	:	+	:	:	:	:		:	:	:	:	:	:
1.3	0.13	0.17	:	:	:	:	:	:		:	:	:	:	:	:
1.7	0.23	0.17	:	:	:	:	:	:		:	:	:	:	:	:
2.1	0.33	0.87	:	:	+	:	:	:		:	:	:	:	:	:
2.7	0.43	0.87	:	:	:	:	:	:		:	:	:	:	:	:
3.4	0.53	1.22	:	:	+	:	:	:		:	:	:	:	:	:
4.3	0.63	1.56	:	:	+	:	:	:		:	:	:	:	:	:
5.4	0.73	4.17	:	:	:	+	:	:		:	:	:	:	:	:
6.8	0.83	5.73	:	:	:	:	+	:		:	:	:	:	:	:
8.5	0.93	7.81	:	:	:	:	+	:		:	:	:	:	:	:
11	1.03	15.45	:	:	:	:	:	+		:	:	:	:	:	:
13	1.13	23.61	:	:	:	:	:	+		:	:	:	:	:	:
17	1.23	51.39	:	:	:	:	:	:		+	:	:	:	:	:
21	1.33	78.47	:	:	:	:	:	:		:	+	:	:	:	:
27	1.43	92.36	:	:	:	:	:	:		:	:	+	:	:	:
34	1.53	97.05	:	:	:	:	:	:		:	:	:	+	:	:
43	1.63	98.78	:	:	:	:	:	:		:	:	:	:	+	:
54	1.73	99.48	:	:	:	:	:	:		:	:	:	:	:	+
68	1.83	99.65	:	:	:	:	:	:		:	:	:	:	:	+
85	1.93	99.65	:	:	:	:	:	:		:	:	:	:	:	+
107	2.03	99.83	:	:	:	:	:	:		:	:	:	:	+	:
135	2.13	99.83	:	:	:	:	:	:		:	:	:	:	:	+
170	2.23	100.00	:	:	:	:	:	:		:	:	:	:	:	+

File B:BRADNER.VGC Threshold/Anomaly Statistics
 COPPER

B:BRADNER.VGC / col 3	Normal	Log-xform	Un-xform
No. Samples	575		
Min	2.00		
Max	95.00		
Mean	13.87	1.07	11.85
Standard Deviation	8.87	0.24	1.75
Threshold	31.62	1.56	36.46
Anomalous	40.49	1.81	63.94
Very Anomalous	49.37	2.05	112.15

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)		
(9)	37.00	* thresh	1.57	* thresh	
(23)	34.00	* thresh	1.53		
(26)	35.00	* thresh	1.54		
(35)	39.00	* thresh	1.59	* thresh	
(43)	44.00	** ANOM	1.64	* thresh	
(44)	74.00	*** V ANOM	1.87	** ANOM	
(124)	35.00	* thresh	1.54		
(163)	35.00	* thresh	1.54		
(224)	37.00	* thresh	1.57	* thresh	
(233)	36.00	* thresh	1.56		
(326)	54.00	*** V ANOM	1.73	* thresh	
(336)	36.00	* thresh	1.56		
(399)	49.00	** ANOM	1.69	* thresh	
(409)	40.00	* thresh	1.60	* thresh	
(429)	35.00	* thresh	1.54		
(461)	36.00	* thresh	1.56		
(531)	95.00	*** V ANOM	1.98	** ANOM	
(542)	35.00	* thresh	1.54		
(559)	32.00	* thresh	1.51		
(567)	72.00	*** V ANOM	1.86	** ANOM	

Class % vs. LOG10(PPM): B:BRADNER.VGC
COPPER

upper			F R E Q U E N C Y (%)				
Log10	Class%	Cum%	1	2	3	4	5
0.03	0.00	0.00	:	:	:	:	:
0.13	0.00	0.00	:	:	:	:	:
0.23	0.00	0.00	:	:	:	:	:
0.33	0.87	0.87	*	:	:	:	:
0.43	0.00	0.87	:	:	:	:	:
0.53	0.87	1.74	*	:	:	:	:
0.63	2.78	4.52	***	:	:	:	:
0.73	5.91	10.43	*****	:	:	:	:
0.83	4.17	14.61	****	:	:	:	:
0.93	6.78	21.39	*****	:	:	:	:
1.03	21.74	43.13	M*****	:	:	:	:
1.13	12.35	55.48	*****	:	:	:	:
1.23	22.26	77.74	*****	:	:	:	:
1.33	8.87	86.61	*****	:	:	:	:
1.43	7.48	94.09	*****	:	:	:	:
1.53	2.61	96.70	T***	:	:	:	:
1.63	2.26	98.96	**	:	:	:	:
1.73	0.35	99.30	A	:	:	:	:
1.83	0.17	99.48	:	:	:	:	:
1.93	0.35	99.83	:	:	:	:	:
2.03	0.17	100.00	V	:	:	:	:

Cumulative Percent vs. Log(PPM): B:BRADNER.VGC
COPPER

upper			P R O B A B I L I T Y (C U M %)												
PPM	Log10	Cum%	.01	.1	.5	2	10	30	50%	70	90	98	99	999	9999
1.7	0.23	0.00	:	:	:	:	:	:		:	:	:	:	:	:
2.1	0.33	0.87	:	:	+	:	:	:		:	:	:	:	:	:
2.7	0.43	0.87	:	:	:	:	:	:		:	:	:	:	:	:
3.4	0.53	1.74	:	:	:	+	:	:		:	:	:	:	:	:
4.3	0.63	4.52	:	:	:	+	:	:		:	:	:	:	:	:
5.4	0.73	10.43	:	:	:	:	+	:		:	:	:	:	:	:
6.8	0.83	14.61	:	:	:	:	+	:		:	:	:	:	:	:
8.5	0.93	21.39	:	:	:	:	:	+		:	:	:	:	:	:
11	1.03	43.13	:	:	:	:	:	:		+	:	:	:	:	:
13	1.13	55.48	:	:	:	:	:	:		+	:	:	:	:	:
17	1.23	77.74	:	:	:	:	:	:		:	+	:	:	:	:
21	1.33	86.61	:	:	:	:	:	:		:	:	+	:	:	:
27	1.43	94.09	:	:	:	:	:	:		:	:	:	+	:	:
34	1.53	96.70	:	:	:	:	:	:		:	:	:	:	+	:
43	1.63	98.96	:	:	:	:	:	:		:	:	:	:	:	+
54	1.73	99.30	:	:	:	:	:	:		:	:	:	:	+	:
68	1.83	99.48	:	:	:	:	:	:		:	:	:	:	+	:
85	1.93	99.83	:	:	:	:	:	:		:	:	:	:	+	:
107	2.03	100.00	:	:	:	:	:	:		:	:	:	:	+	+

File B:BRADNER.VGC Threshold/Anomaly Statistics
MOLYBDENUM

B:BRADNER.VGC / col 2 Normal Log-xform Un-xform

No. Samples	383		
Min	1.00		
Max	20.00		
Mean	2.71	0.37	2.32
Standard Deviation	2.06	0.23	1.69
Threshold	6.84	0.82	6.65
Anomalous	8.91	1.05	11.25
Very Anomalous	10.97	1.28	19.05

Listing of threshold & anomalous values:

Sample	Raw Value		Log10(Value)	
(22)	20.00	*** V ANOM	1.30	*** V ANOM
(23)	8.00	* thresh	0.90	* thresh
(24)	19.00	*** V ANOM	1.28	** ANOM
(44)	7.00	* thresh	0.85	* thresh
(46)	7.00	* thresh	0.85	* thresh
(54)	18.00	*** V ANOM	1.26	** ANOM
(55)	9.00	** ANOM	0.95	* thresh
(107)	7.00	* thresh	0.85	* thresh
(127)	7.00	* thresh	0.85	* thresh
(230)	7.00	* thresh	0.85	* thresh
(298)	7.00	* thresh	0.85	* thresh
(299)	7.00	* thresh	0.85	* thresh
(376)	17.00	*** V ANOM	1.23	** ANOM



986-5211

VANGEOCHEM LAB LTD. 1521 PEMBERTON AVE., NORTH VANCOUVER, B.C., CANADA 604-~~888~~2112

V7P 2S3

Nov. 8 1983

TO: Orequest Consultants
#404 - 595 Howe St.
Vancouver, B C V6C 2T5

FROM: Vangeochem Lab Ltd.
1521 Pemberton Ave.
North Vancouver, B.C. V7P 2S3

SUBJECT: Analytical procedure used to determine hot acid soluble arsenic
in geochemical silt, soil, lake sediments and rock samples.

for geochem soil humus , rock samples

1. Sample Preparation

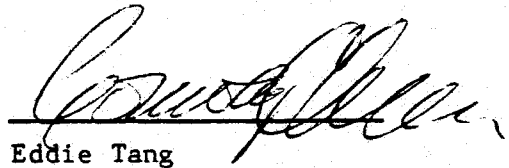
- (a) Geochemical soil, silt, lake sediments or rock samples were received in the laboratory in wet-strength 3½ x 6½ Kraft paper bags and rock samples in 4" x 6" Kraft paper bags.
- (b) The wet samples were dried in a ventilated oven.
- (c) The dried soil and silt samples were sifted by hands using a 8" diameter 80-mesh stainless steel sieves. The plus 80-mesh fraction was rejected and the minus 80-mesh fraction was transferred into a nwq bag for analysis later.
- (d) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Digestion

- (a) 0.25 gram of the minus 80-mesh sample was used. Samples were weighed out by using a top-loading balance.
- (b) Samples were heated in a sand bath with concentrated perchloric acid (70 - 72% HClO₄ by weight) at a medium heat for four hours.
- (c) The digested samples were diluted with demineralized water.

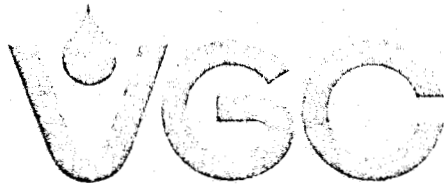
3. Method of Analysis

- (a) Potassium iodide and stannous chloride in HCL were added to the digested samples.
 - (b) Zinc metal was introduced and the arsenic in solution was gassed off as arsene through a glass wool scrubber plug saturated with lead acetate and into a solution of silver diethyldithiocarbamate in chloroform with l-ephedrine, forming a red complex with the silver diethyldithiocarbamate.
 - (c) The concentration of the arsenic was determined colorimetrically by comparing the intensity of the color of the red complex with a set of known standards prepared in a similar fashion as the samples.
4. The analyses were supervised or determined by Mr. Eddie Tang or Mr. Conway Chun and their laboratory staff.



Eddie Tang

VANGEOCHEM LAB LTD.



986-5211

VANGEOCHEM LAB LTD. 1521 PEMBERTON AVE., NORTH VANCOUVER, B.C. CANADA 004-~~XXXXXXXX~~

V7P 2S3

Nov. 8, 1983

To: Orequest Consultants
#404 - 595 Howe St.
Vancouver, B C V6C 2T5

From: Vangeochem Lab Ltd.
1521 Pemberton Avunue
North Vancouver, B.C. V7P 2S3

Subject: Analytical procedure used to determine hot acid soluble
Mo, Cu, Pb, Zn, Ag in geochemical silt, soil and rock samples.

1983 samples

1. Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength $3\frac{1}{2}$ x $6\frac{1}{2}$ Kraft paper bags and rock samples in 4" x 6" Kraft paper bags.
- (b) The wet samples were dried in a ventilated oven.
- (c) The dried soil and silt samples were sifted by hands using a 8" diameter 80-mesh stainless steel sieves. The plus 80-mesh fraction was rejected and the minus 80-mesh fraction was transferred into a new bag for analysis later.
- (d) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Methods of Digestion

- (a) 0.50 gram of the minus 80-mesh samples was used. Samples were weighed out by using a top-loading balance.
- (b) Samples were heated in a sand bath with nitric and perchloric acids (15% to 85% by volume of the concentrated acids respectively).

.....2

(C) The digested samples were diluted with demineralized water to a fixed volume and shaken.

3. Method of Analysis

Mo, Cu, Pb, Zn, Ag analyses were determined by using a Techtron Atomic Absorption Spectrophotometer Model AA4 or Model AA5 with their respective hollow cathode lamps. The digested samples were aspirated directly into an air and acetylene flame, but Mo digestion were aspirated into an acetylene and nitrous flame. The results, in parts per million, were calculated by comparing a set of standards to calibrate the atomic absorption unit and displayed in a strip chart recorder.

4. The analyses were supervised or determined by Mr. Conway Chun or Mr. Eddie Tang and the laboratory staff.

5. Back Ground Correction

A Hydrogen continuum lamp is used to correct the silver ground interferences.


Eddie Tang

VANGEOCHEM LAB LTD.

ET:jl



986-5211

VANGEOCHEM LAB LTD. 1521 PEMBERTON AVE., NORTH VANCOUVER, B.C., CANADA 604-986-2172

V7P 2S3

Nov. 8, 1983

To: Orequest Consultants
#404 - 595 Howe St.
Vancouver, B C V6C 2T5

From: Vangeochem Lab Ltd.
1521 Pemberton Ave.
North Vancouver, B.C. V7P 2S3

Subject: Analytical procedure used to determine Aqua Regia soluble gold
in geochemical samples.

For soil and humus samples

1. Method of Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength 4 x 6 Kraft paper bags or rock samples sometimes in 8" x 12" plastic bags.
- (b) The dried soil and silt samples were sifted by hands using a 8" diameter 80-mesh stainless steel sieve, The plus 80-mesh fraction was rejected and the minus 80-mesh fraction was transferred into a new bag for analysis later.
- (c) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Digestion

- (a) 5.00 - 10.00 grams of the minus 80-mesh samples were used. Samples were weighed out by using a top-loading balance into beakers.
- (b) 20 ml of Aqua Regia (3:1 HCl:HNO₃) were used to digest the samples over a hot plate vigorously.
- (c) The digested samples were filtered and the washed pulps were discarded and the filtrate was reduced to about 5 ml.
- (d) The Au complex ions were extracted into diisobutyl ketone and thiourea medium. (Anion exchange liquids "Aliquot 336").

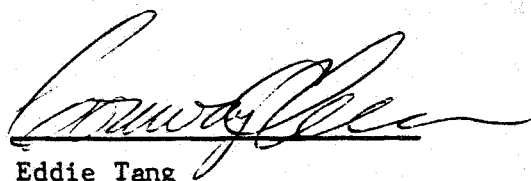
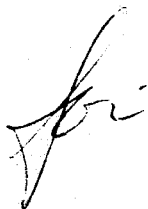
... 2

(e) Separate Funnels were used to separate the organic layer.

3. Method of Detection

The gold analyses were detected by using a Techtron model AA5 Atomic Absorption Spectrophotometer with a gold hollow cathode Lamp. The results were read out on a strip chart recorder. A hydrogen lamp was used to correct any background interferences. The gold values in parts per billion were calculated by comparing them with a set of gold standards.

4. The analyses were supervised or determined by Mr. Conway Chun or Mr. Eddie Tang and his laboratory staff.



Eddie Tang

VANGEOCHEM LAB LTD.

ET: j1



VANGEOCHEM LAB LTD.

1521 PEMBERTON AVE., NORTH VANCOUVER, B.C., CANADA V7P 2S3 (604) 986-5211

Nov. 8 1983

To: Orequest Consultants
#404 - 595 Howe St.
Vancouver, B C V6C 2T5

From: Vangeochem Lab Ltd.
1521 Pemberton Avenue
North Vancouver, B.C. V7P 2S3

Subject: Analytical procedure used to determine gold by fire-assay method and detected by atomic absorption spec. in geological samples.

For samples requested for Fireassays- AAS finished

1. Method of Sample Preparation

- (a) Geochemical soil, silt or rock samples were received in the laboratory in wet-strength 4" x 6" Kraft paper bags or rock samples sometimes in 8" x 12" plastic bags.
- (b) The dried soil and silt samples were sifted by hands using a 8" diameter 80-mesh stainless steel sieve. The plus 80-mesh fraction was rejected and the minus 80-mesh fraction was transferred into a new bag for analysis later.
- (c) The dried rock samples were crushed by using a jaw crusher and pulverized to 100-mesh or finer by using a disc mill. The pulverized samples were then put in a new bag for later analysis.

2. Method of Extraction

- (a) 20.0 - 30.0 grams of the pulp samples were used. Samples were weighed out by using a top-loading balance into a fusion pot.
- (b) A Flux of litharge, soda ash, silica, borax, flour, or potassium nitrite is added, then fused at 1900°F and a lead button is formed.
- (c) The gold is extract by cupellation and part with diluted nitric acid.
- (d) The gold bead is saved or measurement later.



VANGEOCHEM LAB LTD.

1521 PEMBERTON AVE., NORTH VANCOUVER, B.C., CANADA V7P 2S3

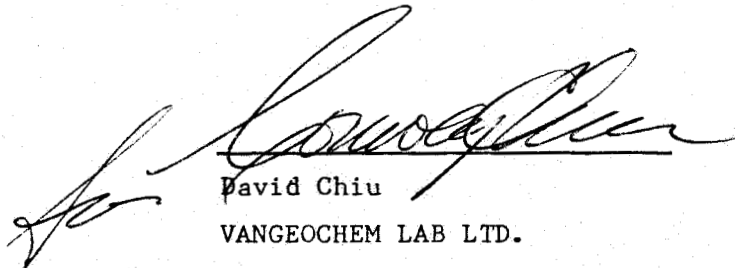
(604) 986-5211

- 2 -

3. Method of Detection

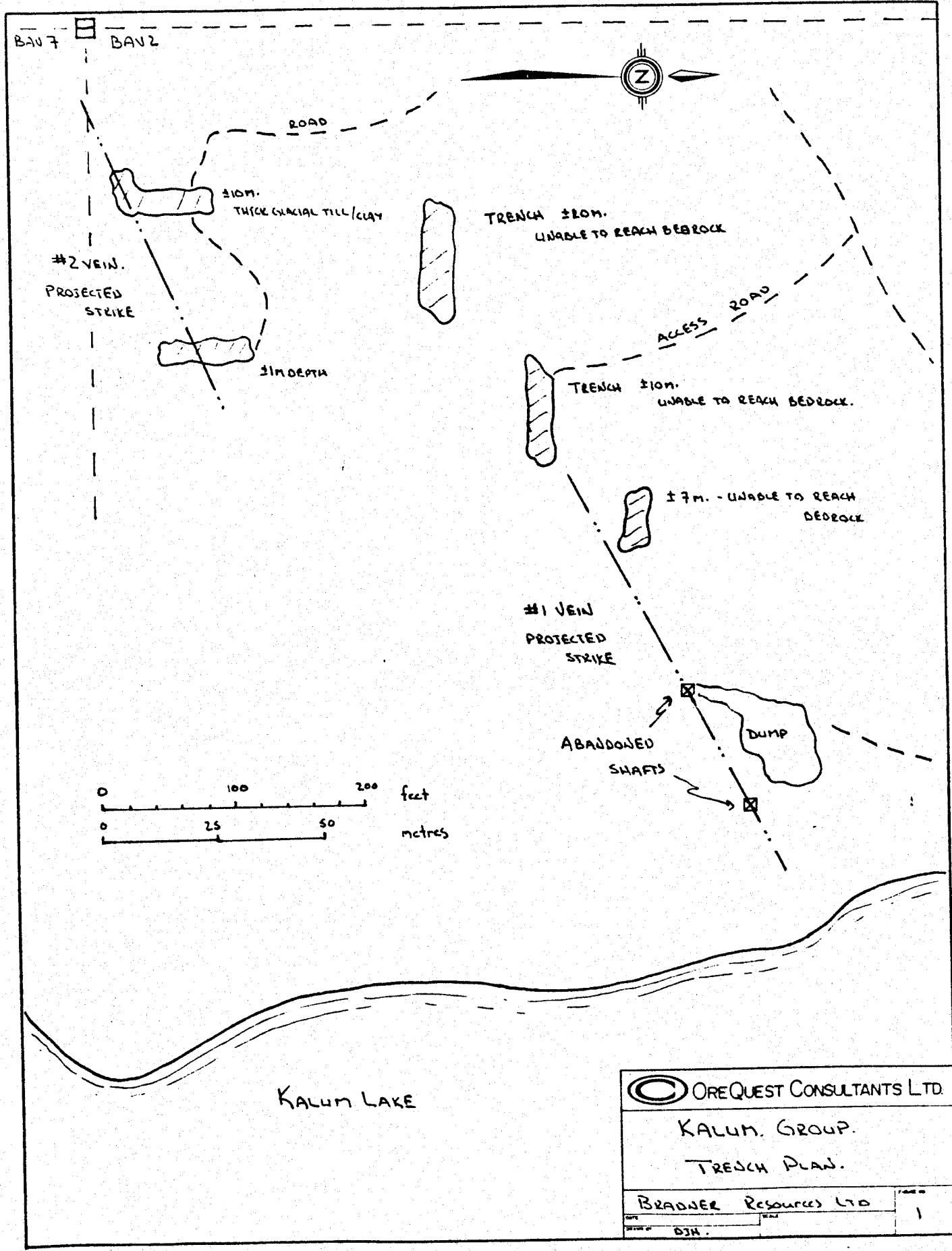
- (a) The gold bead is dissolved by boiling with sodium cyanide, hydrogen peroxide and ammonium hydroxide.
- (b) The gold analyses were detected by using a Techtron model AA5 Atomic Absorption Spectrophotometer with a gold hollow cathode lamp. The results were read out on a strip chart recorder. The gold values in parts per billion were calculated by comparing them with a set of gold standards.


4. The analyses were supervised or determined by Mr. Conway Chun or Mr. David Chiu and his laboratory staff.

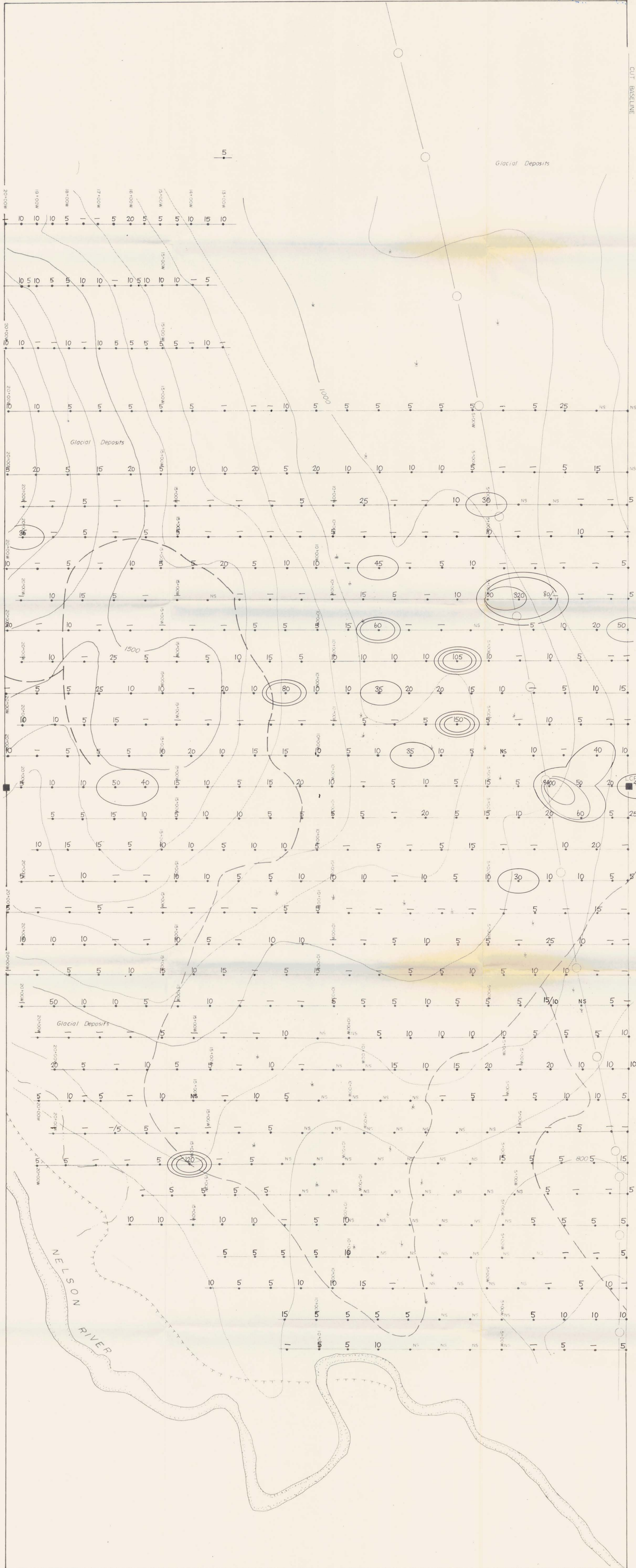


David Chiu
VANGEOCHEM LAB LTD.

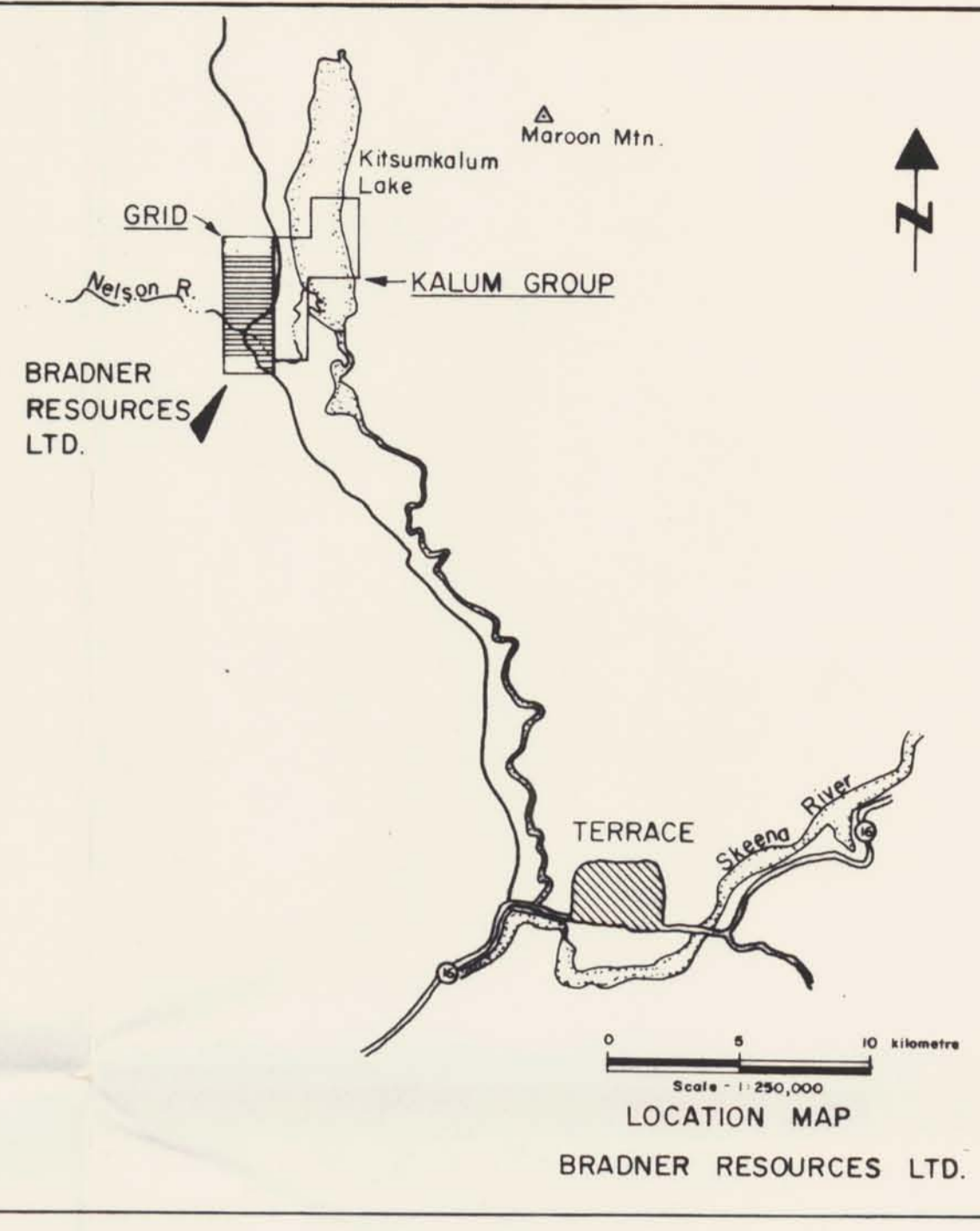
DC:jl



 OREQUEST CONSULTANTS LTD.	
KALUM GROUP.	
TRENCH PLAN.	
BRADNER Resources LTD	
<small>DATE</small> <small>BY</small>	<small>SCALE</small> <small>1</small>
<small>DATE</small> <small>BY</small>	<small>SCALE</small> <small>1</small>



20°00'N
19°00'N
18°00'N
17°00'N
16°00'N
15°00'N
14°00'N
13°00'N
12°00'N
11°00'N
10°00'N
9°00'N
8°00'N
7°00'N
6°00'N
5°00'N
4°00'N
3°00'N
2°00'N
1°00'N
0°00'
1°00'S
2°00'S
3°00'S
4°00'S
5°00'S
6°00'S
7°00'S
8°00'S
9°00'S
10°00'S
11°00'S
12°00'S
13°00'S
14°00'S
15°00'S
16°00'S
17°00'S
18°00'S



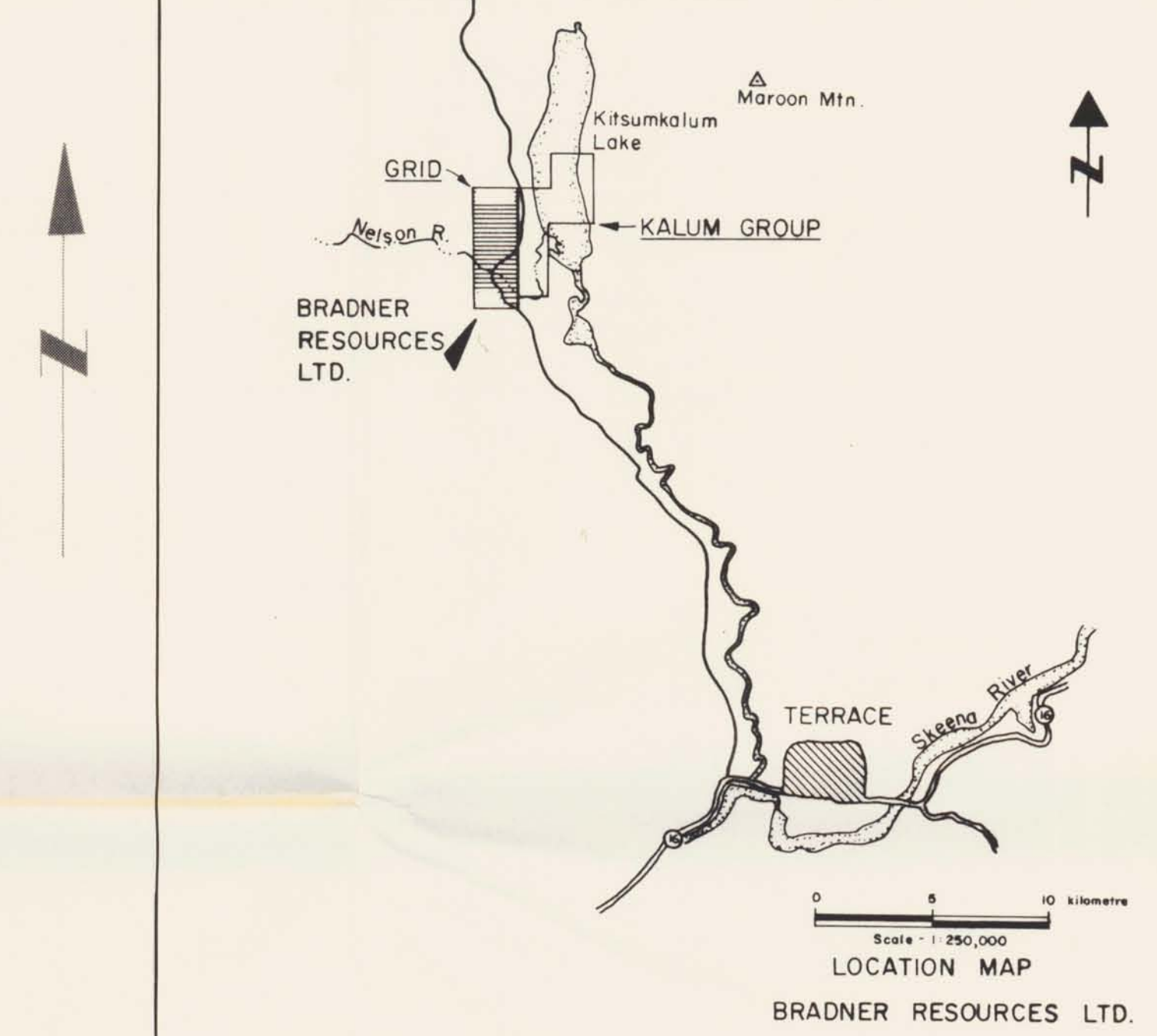
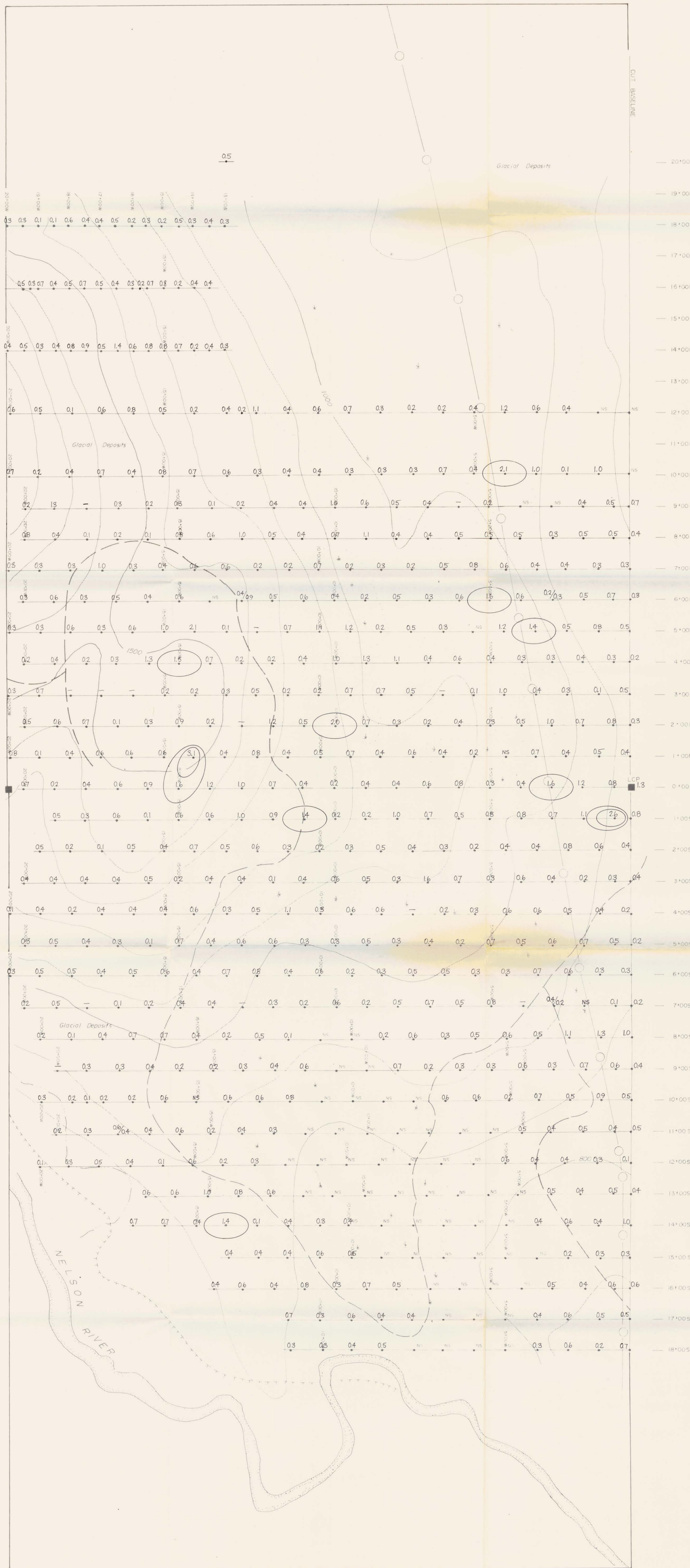
LEGEND

- Very Anomalous ≥ 95 ppb
- Anomalous ≥ 52 ppb
- Threshold ≥ 29 ppb
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- Slope
- ~ Stream
- ⊛ Swamp
- Grid line & Soil sample location
- NS No sample taken

Scale - 1:5,000

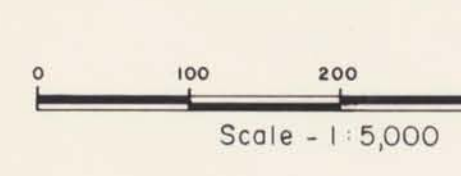
**GEOLOGICAL BRANCH
ASSESSMENT REPORT**
13,303

OREQUEST CONSULTANTS LTD.	
SOIL GEOCHEMISTRY	
Au (ppb)	
Skeena Mining Division, NTS 1031/10 and 15W	
BRADNER RESOURCES LTD.	
DATE NOV/1984	FIGURE NO. 4
DRAWN BY P.Y.	SCALE 1:5,000



LEGEND

- Very Anomalous ≥ 4.6 ppm
- Anomalous ≥ 2.5 ppm
- Threshold ≥ 1.4 ppm
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- Slope
- Stream
- Swamp
- Grid line & Soil sample location
- No sample taken



13,303

OREQUEST CONSULTANTS LTD.

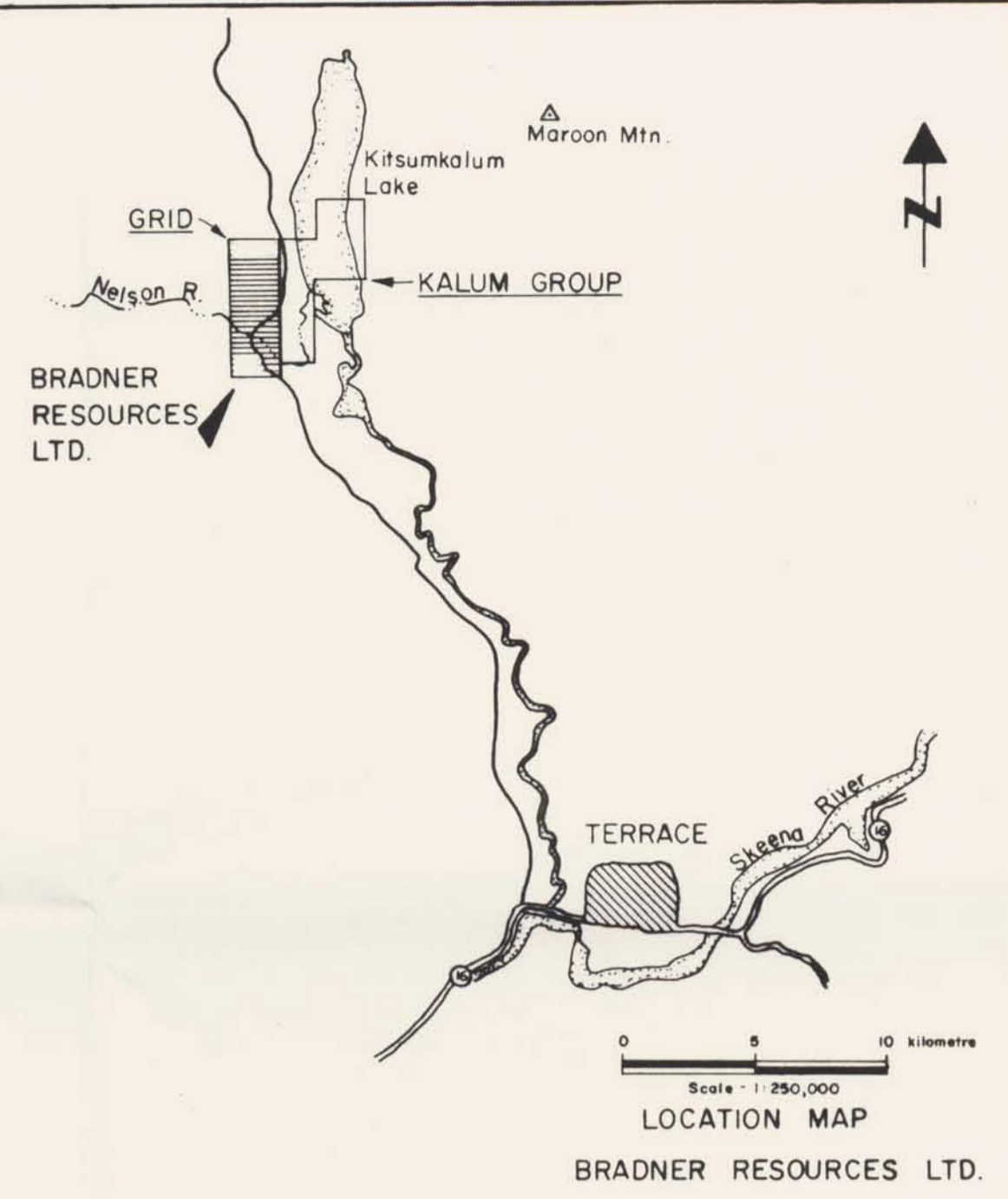
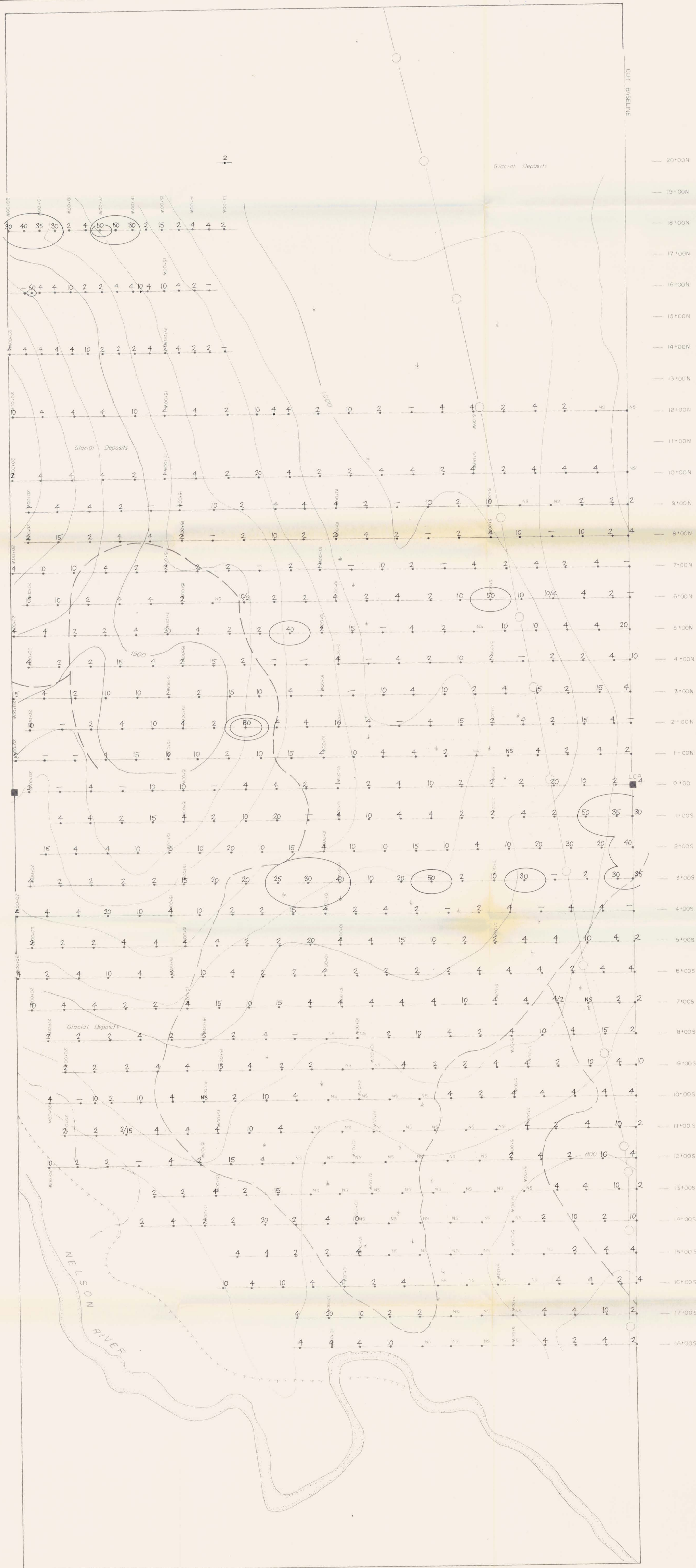
SOIL GEOCHEMISTRY

Ag (ppm)

Skeena Mining Division, NTS 1031/10 and 15W

BRADNER RESOURCES LTD.

DATE	NOV/1984	SCALE	1:5,000	FIGURE NO.	5
DRAWN BY	BY				



LEGEND

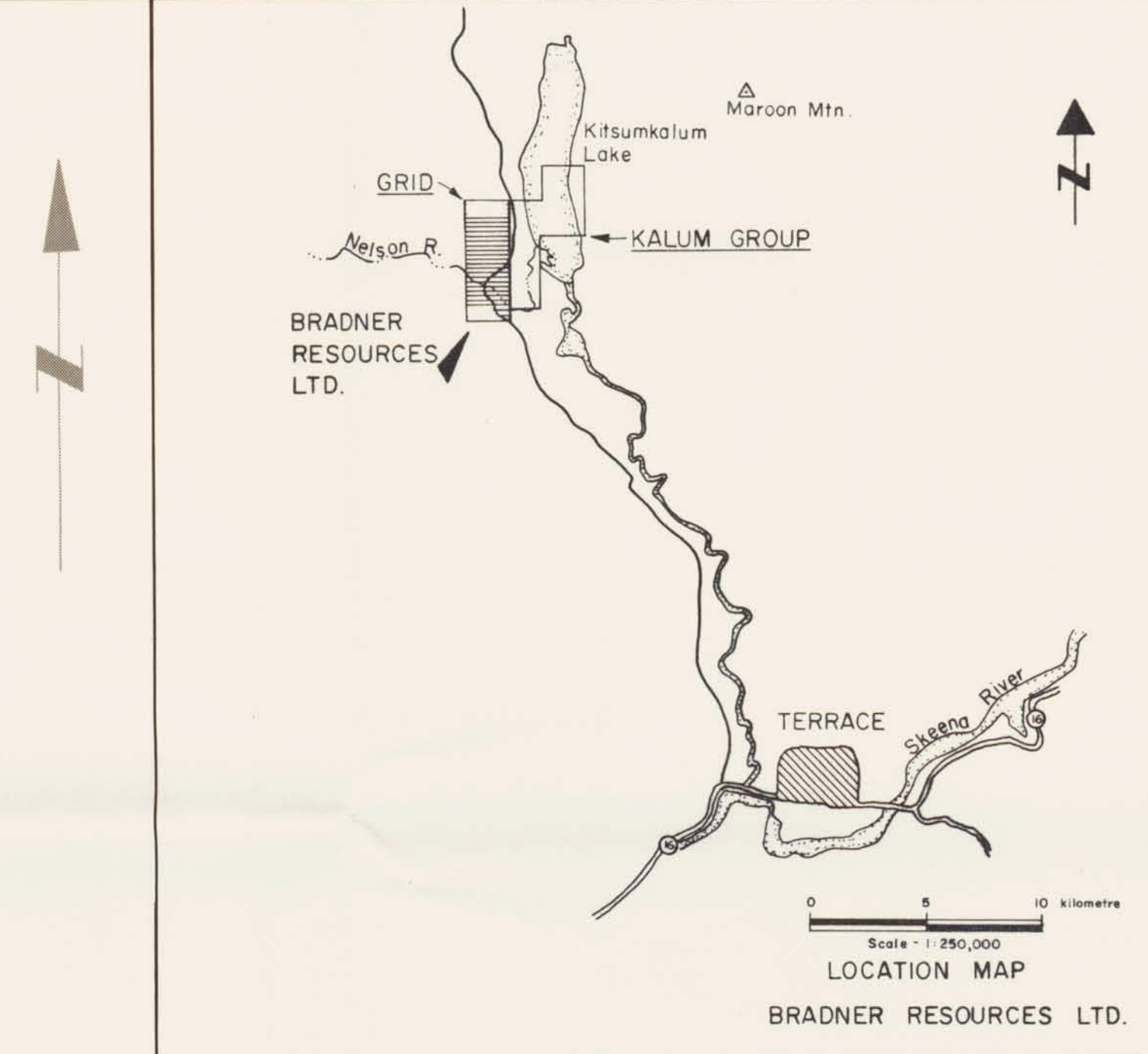
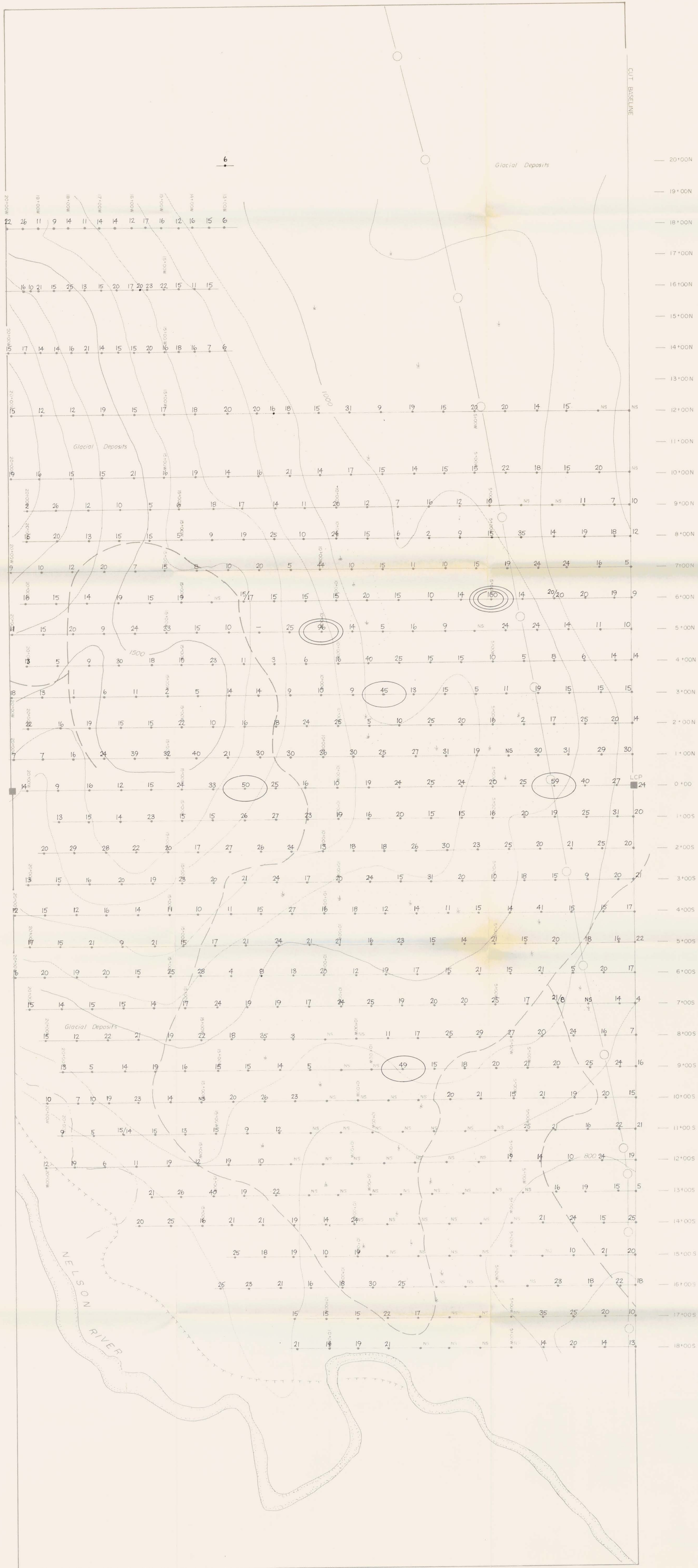
- Very Anomalous ≥ 121 ppm
- Anomalous ≥ 53 ppm
- Threshold ≥ 23 ppm
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- ▴ Slope
- Stream
- Swamp
- Grid line & Soil sample location
- No sample taken

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

13,303

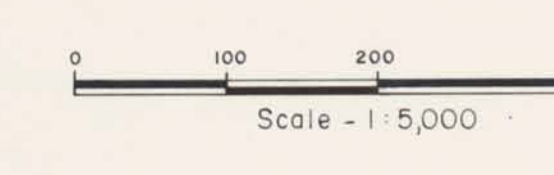
Scale - 1:5,000

OREQUEST CONSULTANTS LTD.	
SOIL GEOCHEMISTRY As(ppm)	
Skeena Mining Division, NTS 1031/10 and 15W	
BRADNER RESOURCES LTD.	
DATE NOV/1984	FIGURE NO. 6
DRAWN BY P.Y.	SCALE 1:5,000



LEGEND

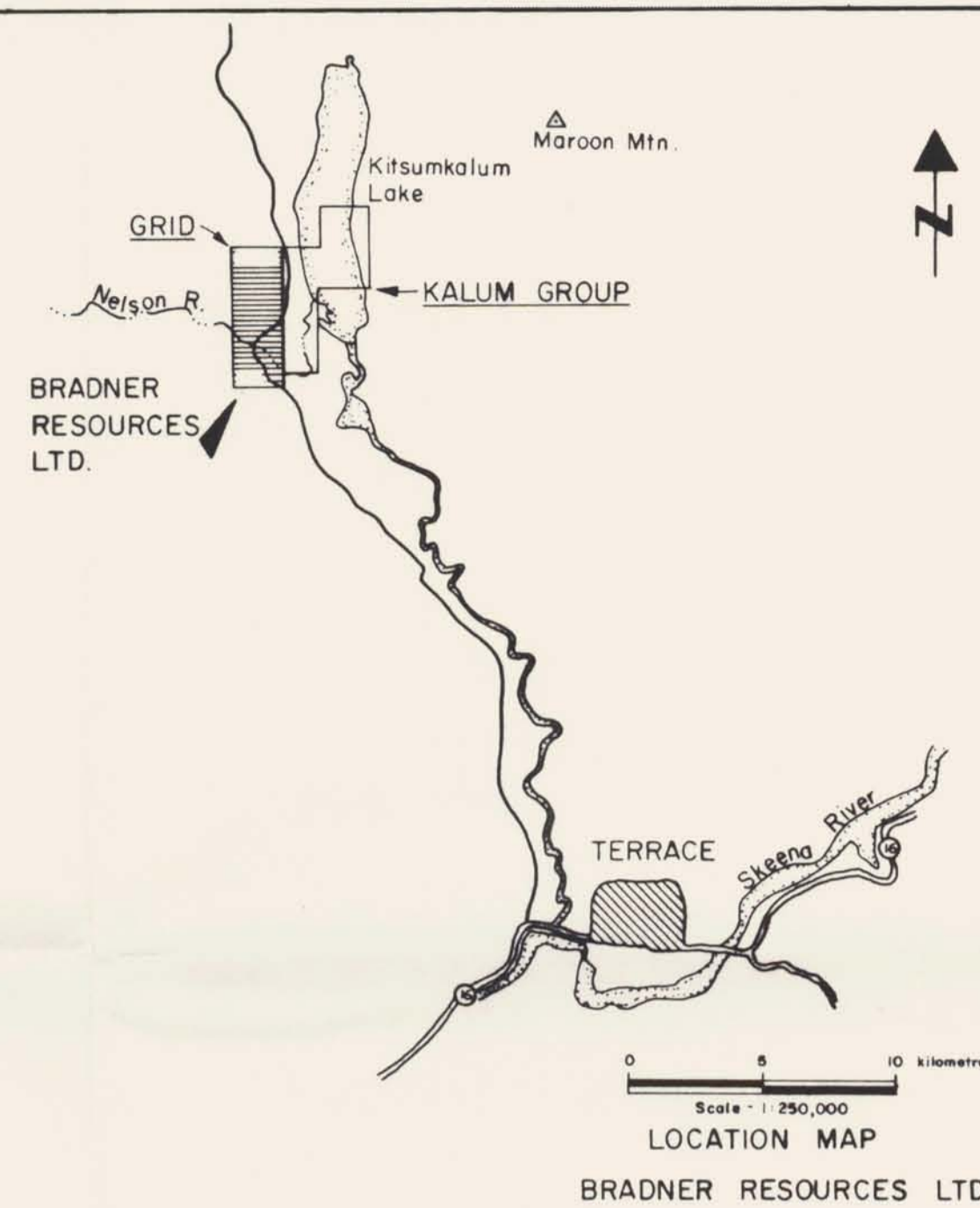
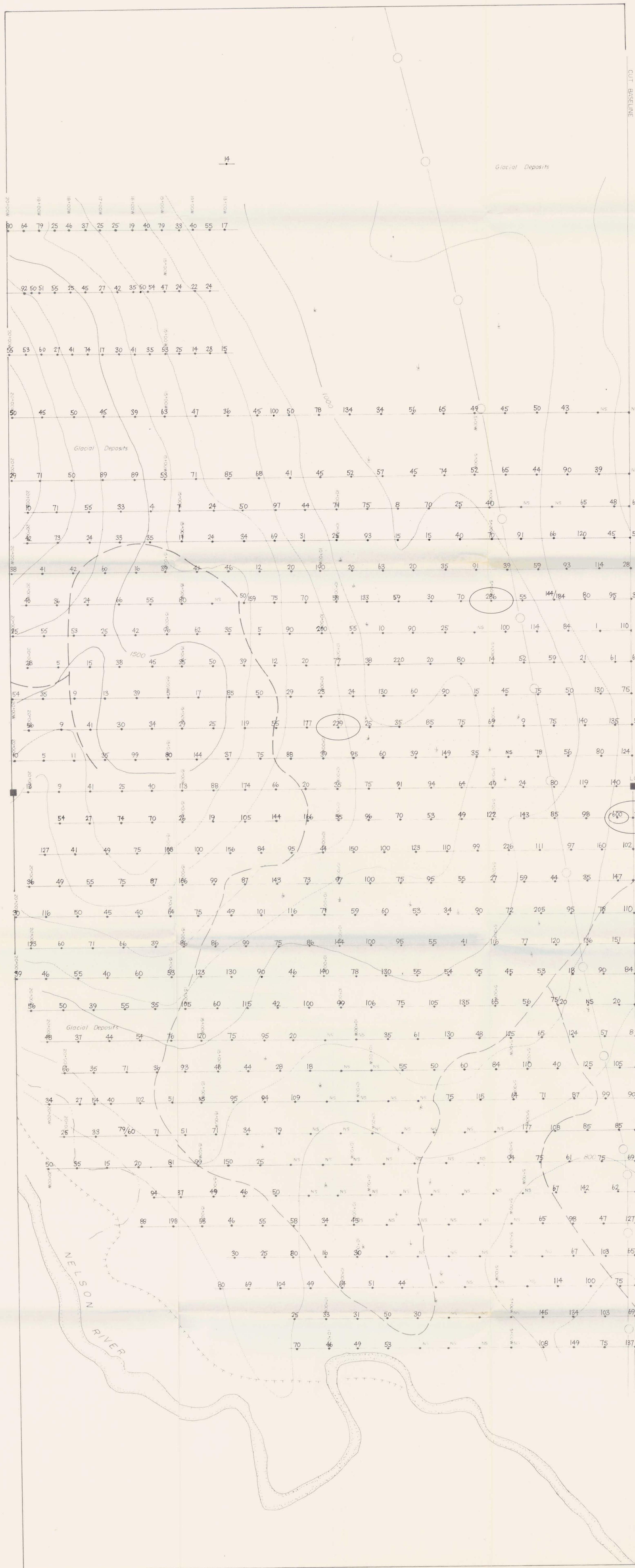
- Very Anomalous ≥ 114 ppm
- Anomalous ≥ 70 ppm
- Threshold ≥ 43 ppm
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- Slope
- Stream
- Swamp
- Grid line & Soil sample location
- No sample taken



**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

13,303

OREQUEST CONSULTANTS LTD.	
SOIL GEOCHEMISTRY	
Pb (ppm)	
Skeena Mining Division, NTS 1031/10 and 15W	
BRADNER RESOURCES LTD.	
DATE NOV/1984	SCALE 1:5,000
DRAWN BY P.Y.	PAGE NO. 7



LEGEND

- Very Anomalous ≥ 909 ppm
- Anomalous ≥ 449 ppm
- Threshold ≥ 222 ppm
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- ~ Slope
- Stream
- ⊕ Swamp
- Grid line & Soil sample location
- NS No sample taken

Scale - 1:5,000

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**
13,303

OREQUEST CONSULTANTS LTD.

SOIL GEOCHEMISTRY
Zn (ppm)

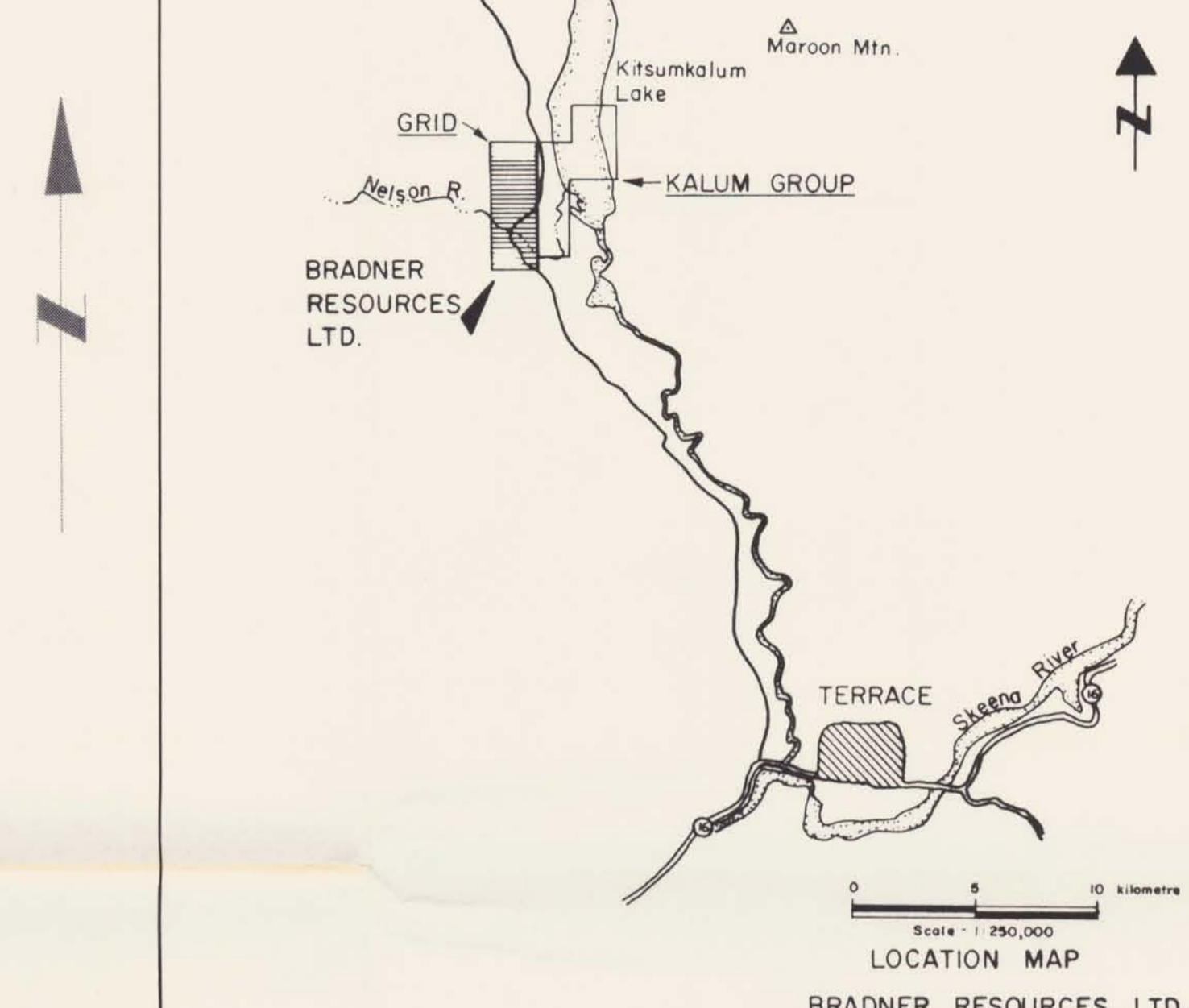
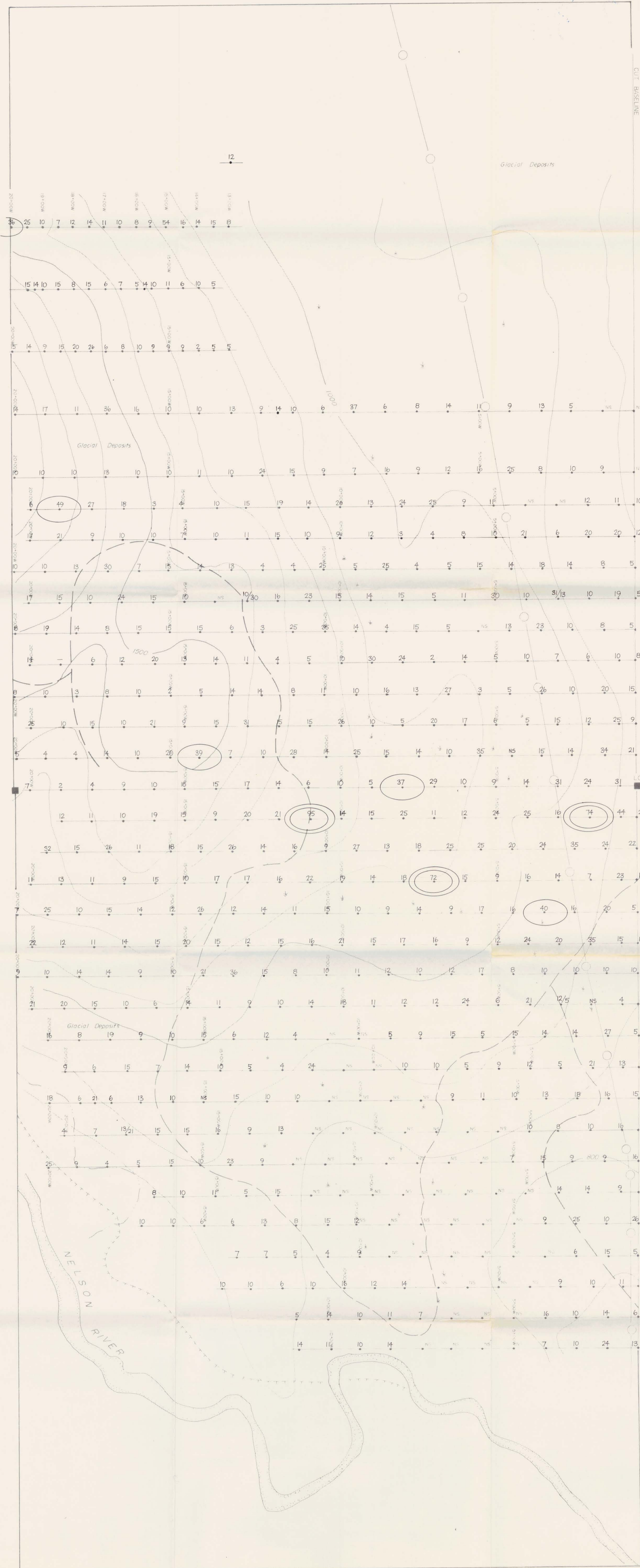
Skeena Mining Division, NTS 1031/10 and 15W

BRADNER RESOURCES LTD.

DATE: NOV/1984

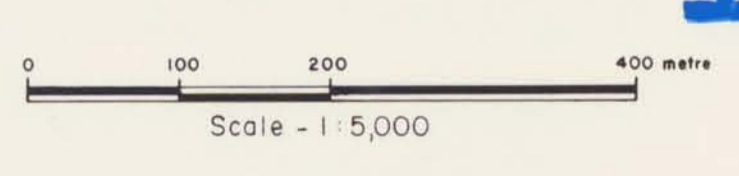
SCALE: 1:5,000

8



LEGEND

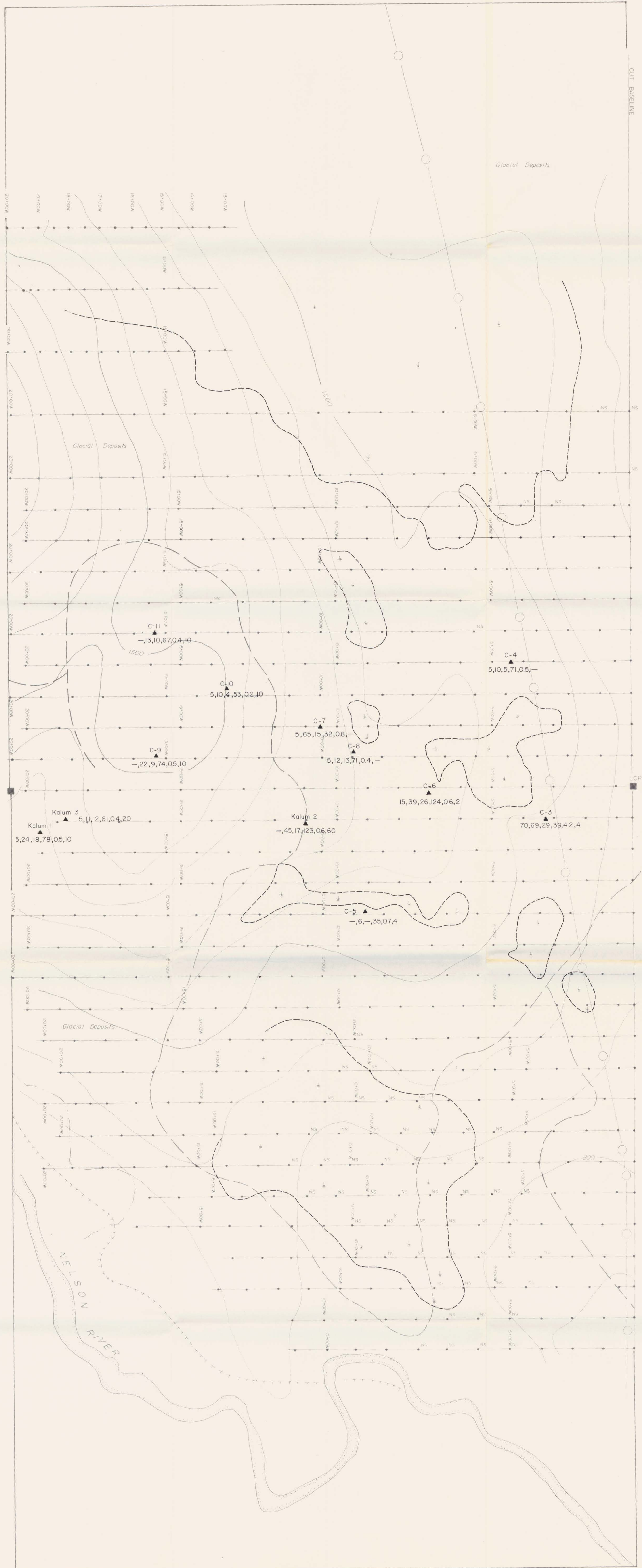
- Very Anomalous ≥ 112 ppm
- Anomalous ≥ 64 ppm
- Threshold ≥ 36 ppm
- Claim post & Claim outline
- Gravel Road
- B.C. Hydro & Transmission line
- Slope
- Stream
- Swamp
- Grid line & Soil sample location
- NS No sample taken



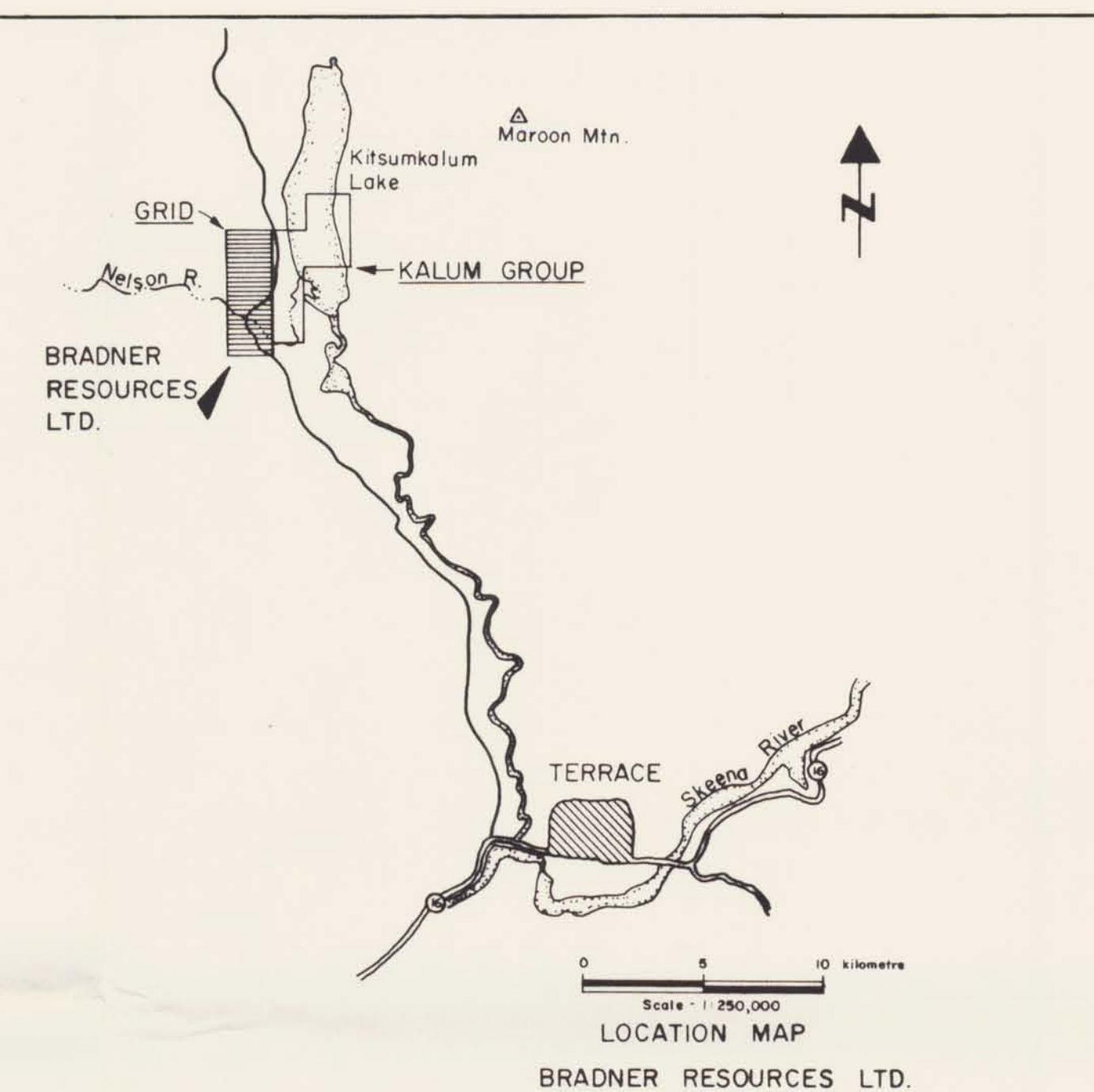
**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

13,303

OREQUEST CONSULTANTS LTD.		
SOIL GEOCHEMISTRY		
Cu (ppm)		
<small>Skeena Mining Division, NTS 1031/10 and 15W</small>		
BRADNER RESOURCES LTD.		<small>FIGURE NO.</small>
<small>DATE</small> NOV / 1984	<small>SCALE</small> 1:5,000	9
<small>DRAWN BY</small> PY		



20°00'N
19°00'N
18°00'N
17°00'N
16°00'N
15°00'N
14°00'N
13°00'N
12°00'N
11°00'N
10°00'N
9°00'N
8°00'N
7°00'N
6°00'N
5°00'N
4°00'N
3°00'N
2°00'N
1°00'N
0°00'
1°00'S
2°00'S
3°00'S
4°00'S
5°00'S
6°00'S
7°00'S
8°00'S
9°00'S
10°00'S
11°00'S
12°00'S
13°00'S
14°00'S
15°00'S
16°00'S
17°00'S
18°00'S



- LEGEND**
- Claim post & Claim outline
 - Gravel Road
 - B.C. Hydro & Transmission line
 - ~ Slope
 - ~ Stream
 - ⊙ Swamp
 - Grid line & Soil sample location
 - NS No sample taken
 - ▲ Rock sample Au(ppb), Cu, Pb, Zn, Ag, As (ppm)

Scale - 1:5,000

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

13,303

OREQUEST CONSULTANTS LTD.

ROCK SAMPLE LOCATION

Skeena Mining Division, NTS 1031/10 and 15W
BRADNER RESOURCES LTD.

DATE: NOV/1984 SCALE: 1:5,000 REGION NO: 10