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GEOLOGICAL and GEOCHEMICAL REPORT

GRANITE MINERAL CLAIM

OMINECA MINING DIVISION

N.T.S. 94C/5W ✓

56°28' 125°52'

RLC	ED
DEC 8	1993
Gold Division	Office
VANCOUVER B.C.	

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GEOLOGICAL BRANCH  
ASSESSMENT REPORT

23,150

**GEOLOGICAL and GEOCHEMICAL REPORT**

**GRANITE MINERAL CLAIM**

GRANITE M.C 20 units Record No. 12642 Tenure No. 242792

located in the Omineca Mining Division  
of British Columbia

N.T.S 94C/5W

Latitude 56 Degrees 28 Minutes North

Longitude 125 degrees 55 minutes West

Work applied to: Granite Mineral Claim

Operator

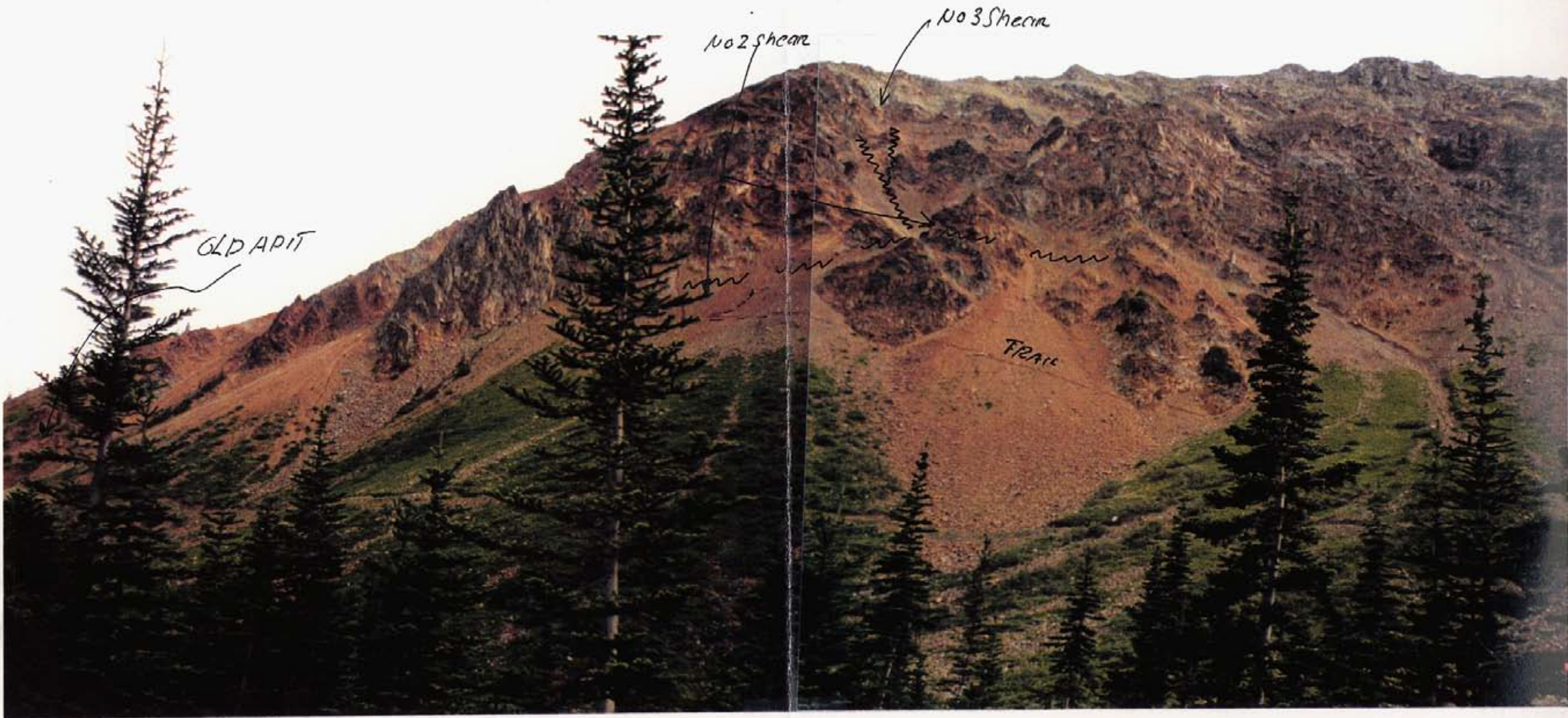
CANASIL RESOURCES INC.  
1695 Marine Drive  
North Vancouver, B. C. V7P 1V1

Prepared by: \_\_\_\_\_

*P.J. Weishaupt*  
Paul J. Weishaupt F.M.C 128530

Date submitted: \_\_\_\_\_

*December 8, 1993*



OLD APIT

No 2 Shear

No 3 Shear

TRAIL

⊙ CAMP

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## 1. INTRODUCTION

The Granite M.C. Record No. 12842, Tenure No. 242792 consists of twenty units. The area was staked by Paul Weishaupt in 1990. The claim covers the ground where extensive exploration was done in 1936 and 1937 by the Consolidate Mining and Smelting Company (Cominco), who located gold and silver mineralization on the property.

Various other companies explored the area during 1974 to 1981 but failed to identify gold or silver mineralization of economic significance and the interpretation of results were inconclusive.

Since 1990 some modest exploration work was conducted by Paul Weishaupt and by Teck Exploration Ltd. in conjunction with their Porphyry Creek Project.

The purpose of the 1993 exploration program was to locate, trench and sample some of the gold bearing structures as reported in the early reports by the Consolidated Mining and Smelting Company.

## 2. LOCATION, ACCESS and PHYSIOGRAPHIC SETTING

The Granite Mineral Claim is located at Latitude 56 degrees 28 minutes North and Longitude 125 degrees 55 minutes West in the Aiken Lake District of the Omineca Mining Division of British Columbia.

The Omineca Resource Access Road (O.M.A.R) and all weather mainline logging roads provide access to within 3.6 Km. of the property. This last 3,6 Km. is stricly a 4 wheel drive road. The distance from Ft. St. James via Germansen Landing to the junction of the 4 wheel drive road is 365 Km. From Prince George via Windy Point to the junction is approx. 438 Km.

The claim covers the south wall of a north-easterly facing cirque draining into Lay Creek. Topography is very steep except for local areas along the creeks. Elevations range from 1300 meters at the north-east corner to 2070 meters at the south-west corner of the property. The area of the gold-silver mineralization is above 1500 meters with only minor vegetation consisting of mountain willow and fireweed.

Location, Access and Physiographic setting continued-

At lower elevation the vegetation comprises a mix of sub-alpine lodge pole pine and spruce. The climate is generally moderate with temperatures ranging from +30 to -30 celcius, typical of the interior of the province with precipitation of 900 mm. per year. Ample water is available for diamond drilling and mine development. The exploration season can begin around June 1 and continue to the end of September. Snow remains on the north facing cirque until July.

3. LAND STATUS

The Granite Mineral Claim, record No. 12642, tenure No. 242792 consists of 20 units.

Paul J. Weishaupt is the recorded owner of the property. Canasil Resources Inc. has an option to acquire 100% of the Granite claim, subject to a 2% N.S.R., by spending \$1,000,000 over 4 years.

A Statement of Work was filed on October 6, 1993.

CLAIM	UNITS	RECORD NO.	TENURE NO.	NEW EXPIRY DATE
Granite	20	12642	242792	8.10.98

#### 4. HISTORY

In 1935 interesting gold assays were obtained by Cominco from ore floats located along a steep, craggy slope. The area was staked in 1936 and 1,142 linear feet of hand trenching was done. 110 feet of drifting in talus was done without reaching bedrock. The trenches were surveyed and sampled at 5 ft. widths. In 1937 a new drift, at somewhat higher elevation, was driven to intersect the surface values obtained in Trench 1+12 No. 1 zone. This surface trench, 65 feet in length, had two sections with gold values - Section 1, 20-30 feet (10 ft. of 0.11 oz/ton gold) and Section 2, 45-65 ft. (20 ft. 0.355 oz/ton gold) Trench Section 1 was intersected underground in the No.1 Cross-cut South. Cross-cut No.2 was too short to intersect Trench Section 2. A proposed 970 ft. Cross-cut to be driven in 1938 was cancelled when all Cominco's efforts were concentrated at the Pinchi Lake Mercury claims staked May 1st. of that year.

In 1940 Douglas Lay, of the Department of Mines, visited the Granite Basin Property, took samples and wrote a report on the property. ( Bulletin 1, 1940 by D. Lay ). The property was idle until 1962 when Emil Bronlund prospected the area and located new showings to the West of the 1937 Adit Workings and consequently staked the Lay Group of Mineral Claims.

In 1963 W. Sinola, geologist for Kerr Addison Gold Mines Ltd., took some samples from the area one of which assayed 0.22 oz/ton gold and 7.9 oz./ton silver. Nothing further was done on the Property until 1971 when D. Stelling staked the Susie Claims. In 1972 and 1973 the claims were under option to Union Miniere Exploration who conducted a soil geochemical survey and 65 ft. of exposure was sampled.

In 1974 Stelling optioned the Susie group to Susie Gold Mines who conducted geochemical soil and rock chip surveys over all the claims. Road access was constructed and trenching to the South-east of the original showings was done. This was a 2 year program.

The showings were re-staked by N. Burmeister in 1979 as the Granite Basin 1-6 Claims and he optioned them to Mark V Petroleum Ltd. early in 1980. Taiga Consulting of Calgary conducted EM and Magnetometer surveys and took chip samples along the road. Nothing more was done until Paul Weishaupt staked the GB 1-6 in October of 1990.

Since 1990 a soil survey was conducted and 123 samples were collected. Trenches were blasted into a cliff face to permit samples to be taken. This survey indicates the presence of gold bearing rocks on the north facing slopes of the cirque. The rock sampling program west of the underground workings confirmed the presence of gold - silver mineralization. 1.0 meters at 0.20 oz\ton gold and 1.78 oz\ton silver and 3 meters at 0.68oz\ton gold and 10.61 oz\ton silver.

5. PROPERTY GEOLOGY

The claims are underlain by Takla group andesite and intercalated sedimentary rocks, invaded by small bodies of Omineca Intrusions. The predominant rocks in the immediate vicinity of the main workings are moderately dark, grey-green porphyritic andesites, with small black hornblend and scattered grey feldspar phenocryst. A few beds of tuff, argillite and impure limestone are intercalated with the andesite. The rock is cut by a grey to greenish-grey "diorite porphyry" with hornblend phenocryst in a fine grained matrix. This porphyry closely resembles the andesite.

The andesite and the "diorite porphyry" are intruded by a medium to light grey to buff coloured, medium grained, sugary "porphyritic diorite" with abundant feldspar phenocryst.

The andesite, the "diorite porphyry" and the "porphyritic diorite" are all cut by well defined dykes, 10 to 100 feet wide, of light grey feldspar porphyry.

The andesite and the "diorite Porphyry" are generally sparsely mineralized with fine grained pyrite but may be well mineralized where they are in contact with the "porphyritic diorite" which is heavily, though somewhat irregularly, pyritized.

Four pyritized bands are exposed within a horizontal distance of about 600 meters between elevations of 1600 to 2000 meters above mean sea level. They appear to consist mainly of sill-like bodies of porphyritic diorite trending about parallel with the bedding of the tuffs and argillites which are well exposed further West on the cirque wall, where they strike North-east and dip 40 to 60 degrees North-west. The most Easterly of these bands is split by an unmineralized porphyry dyke about 18 meters wide producing, at the crest of the ridge, five pyritized bands which have been numbered 1 to 4 consecutively from East to West. In the earlier days most of the work was concentrated on the pyritized bands.

Recent prospecting and mapping indicates that the gold and silver values are associated with a pattern of shearing which cuts across all rock types and contacts and is later than the pyritic zones and probably the latest structural event.

Three zones of shearing, mineralized with gold and silver have been located to date, varying from 1.5 to 11.0 meters in width. Two of the zones have an east-west strike with steep northerly dips and trend parallel to the basin valley axis. One shearing strikes north-south with steep westerly dip.



Property geology continued

No.1 Shear Zone is the most easterly on which the tunnel was driven in 1937. A sample taken by Douglas Lay of Dept.of Mines, across 12 meters in the underground working assayed: gold 0.2 opt.(1940). Little is known of any strike length because of talus cover in both strike directions.

No.2 Shear Zone outcrops at a point 180 meters southwest of the tunnel portal at 85 meter higher elevation. To the east of this outcrop are steep, inaccessible cliffs, to the west the zone is covered by talus for 79 meters and then outcrops on a ridge where it crosses pyritized bands of porphyritic diorite.

No.3 Shear Zone lies 140 meters further southwest at an elevation of 182 meters above the tunnel.

**6. MINERALIZATION**

Rock specimens collected from the three shearzones show a white to light blueish coloured aphanitic groundmass with thin, closely spaced, ribbon like wavy bands of pyrite, patches of carbonates, some vugs and cross fractures. Thin sections show a schistose, chloritic and hydromicas with some carbonates. Two generations of pyrite is evident, an early fine-grained variety, randomly dispersed and in places as a film along shearplanes. The late pyrite is coarser grained, darker yellow and occurs as irregular pods in small cross fractures and as beaded ribbons along the foliations, generally accompanied by very fine-grained tetrahedrite, galena and minor chalcopyrite which appear to be the latest sulphides.

Much of the gold is associated with the basemetal sulphides but some can be seen as extremely fine, dust like, particles in the siliceous matrix. The silver values are all with tetrahedrite.

## 7. 1993 TRENCHING AND ROCK SAMPLING PROGRAM

The objective of the 1993 program was to locate and verify the values reported in 1936-1937 and explain the discrepancies between those results and the ones obtained from 1974 to 1981.

All exposed outcrops were drilled and blasted to ensure correct sampling.

A total of 87 rock samples were taken from five main trenches. All samples were submitted for geochemical analysis to Acme Analytical Laboratories Ltd. of Vancouver, for 31 element analysis using ICP. technique. Gold values in ppb. were determined by atomic absorption.

The following approach has been used to evaluate the results of the rock samples.

<u>BACKGROUND</u>	<u>ABOVE THRESHOLD</u>	<u>WEAKLY ANOMALOUS</u>	<u>ANOMALOUS</u>
Gold - 0-30 ppb	30-100 ppb	100-200 ppb	+200ppb
Silver- 0-1 ppm	1-2 ppm	2-3 ppm	+ 3ppm

Based on the above numerical criteria the 87 rock samples can be separated into the following categories:-

<u>BACKGROUND</u>	<u>ABOVE THRESHOLD</u>	<u>WEAKLY ANOMALOUS</u>	<u>ANOMALOUS</u>
Gold 11 (12.6%)	7 (8.1%)	2 (2.3%)	67 (77%)
Silver 10 (11.5%)	6 (6.9%)	1 (1.2%)	70 (80%)

77% of the samples taken have anomalous values in gold from 201 ppb to 43,800 ppb. 80% have anomalous values in silver from 3ppm to 546.7 ppm.

Some of the significant sections are as follows:

<u>No.2 Shearzone</u>	<u>Sample No's</u>	<u>Width</u>	<u>Au.gr\t</u>	<u>Ag.gr\t</u>
Trench area 1	24087-24090	4.0m	5.0	41.6
	24091-24093	3.0m	4.64	82.4
	24052	1.2m	19.4	546.7
Trench area 2	201334	1.2m	6.68	222.1
	201335	2.0m	1.45	26.1
	201336	0.5m	43.8	213.1
Trench area 3	24096	0.7m	19.5	151.4
<u>No.3 Shearzone</u>				
Trench area 4	24097-24100			
	201301-302	5.0m	5.5	29.5
Trench area 5	201314-324	11.0m	1.8	9.8
	201303-309	7.0m	1.8	10.6
	201310-12	3.0m	3.1	18.7
	201331-32	2.0m	4.7	22.8

Discussion of results

It is extremely difficult to recognise the zones in outcrops as the intrusive rocks as a whole contain in part abundant pyrite and shearing of all rock types is common. Discrepancies between sample results obtained from the zones from 1937-1981 suggests that the sampling was not carried out on the same zones. The base metal mineralization is sparse and often difficult to recognise in the field. Due to the rugged topography access to the zones is often difficult and work progress is slow.

Trenching done on the No.1 Shear Zone, close to the 1937 tunnel failed to expose the old portal due to continuous sloughing of the fine talus from higher elevation. Numerous test pits were dug by hand along the possible strike of the No. 2 Shear Zone, in talus, but no bedrock was reached.

The results of the geochemical analysis strongly indicate elevated gold and silver values over a considerable area with some highgrade pockets.

### SUMMARY AND CONCLUSION

The property has been well documented over the past 50 years. The original work done for Cominco in 1936 and 1937 was concentrated on the No.1 Shearzone and consisted of trenching and underground development. According to Douglas Lay in B.C Dept. of Mines Bulletin No.1 a sample taken by himself across 40 feet in the underground workings, assayed Gold 0.2 oz\ton. The 1993 trenching failed to expose the old underground workings or the west end of the old trenches due to the deep talus, so very little information was obtained from No. 1 Shear Zone

The No 2 Shear Zone, approximately 180 meters southwest of the underground working, is exposed for only a short distance. To the east of the outcrop are steep, inaccessible bluffs, to the west the zone is covered by talus for 79 meters and then outcrops on top of a steep ridge and also on the west face of the ridge approximately 120 meters west of the east exposure. The samples obtained from the No.2 Shear Zone clearly indicate the presence of gold-silver bearing solution in the system, striking east-west with a steep northerly dip. Further exploration work will be required to substantiate if the three outcrops carrying gold values are part of the same zone. Trenching in the talus slope between the outcrops failed to reach bedrock.

The No.3 Shear Zone is located approximately 140 meters southwest and 100 meters higher elevation than the No.2 Shear Zone. Trenching strongly indicates two parallel Shear Zones striking due north to N20 degrees west with a westerly dip of 75-80 degrees. Both Shears are cut off by a fault to the south, striking approximately east-west with a dip of 75 degrees to the south. To the north the shears are covered by talus.

Trenching and rock sampling of the Shears has confirmed the presence of gold-silver bearing systems on the Granite property.

### RECOMMENDATIONS

A two stage exploration program is recommended. The first stage to include detailed grid layout, geological mapping, further trenching and rock sampling of shears, cleanout of the adit portal to sample the underground workings. This should enable the confirmation of the results obtained in 1940.

Contingent on the first stage, the second stage would be Diamond Drilling.

APPENDIX 1  
METHOD OF ANALYSIS

ACME ANALYTICAL LABORATORIES LTD. VANCOUVER, B. C

GEOCHEMICAL LABORATORY METHOD

SAMPLE PREPARATION

Rock samples

The rock samples are dried, crushed and powderized to minus 100 mesh and a 200 gram sample is obtained.

GEOCHEMICAL ANALYSIS (ICP)

0.5 gram samples are digested in hot dilute aqua regia in a boiling water bath and diluted to 10 ml. with demineralized water.

Extracted metals are determined by:

1. ICP - 0.50 gram sample is digested with 3 ml. of 3:1:2 HCL-HNO<sub>3</sub>-H<sub>2</sub>O at 95 degrees celsius for 1 hour and is diluted to 10 ml. with water.

Cu., Pb., Zn. and Ag. are determined by ICP.

GEOCHEMICAL ANALYSIS (AA)

2. 10.0 gram samples that have been ignited overnite at 600 degrees celsius are digested with hot dilute aqua regia and the clear solution obtained is extracted with Methyl Isobutyl Ketone.

Au. is determined in the MIBK extract by Atomic Absorption.

3. Fire assay for Au. and Ag. 1 A.T Sample

The results for Cu., Pb., Zn. and Ag. are reported in Parts Per Million (ppm)

The results for Au. are reported in Parts Per Billion (ppb)

The results for Au. and Ag. Fire Assay are reported in oz\ton.

**APPENDIX 2**

**ASSAY CERTIFICATES FOR ROCK SAMPLES**



## GEOCHEMICAL ANALYSIS CERTIFICATE

Canasil Resources Inc. File # 93-2649

GRANITE Project



1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: Paul Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B %	Al %	Na %	K %	W ppm	Au* ppb
E 201303	4	13	63	25	8.1	16	10	36	5.54	146	<5	<2	<2	26	<.2	4	<2	5	.02	.010	<2	7	.04	33	.13	4	.33	.01	.17	3	2440
E 201304	3	12	46	16	9.5	20	11	22	5.46	144	<5	<2	<2	14	.3	4	<2	5	.02	.009	<2	8	.02	23	.14	4	.28	.01	.16	3	2100
E 201305	4	10	29	13	9.2	14	9	40	4.94	129	<5	<2	<2	6	.4	4	<2	4	.01	.009	<2	14	.02	23	.13	<2	.24	.01	.16	2	1020
E 201306	4	15	52	13	16.2	12	8	51	3.61	115	<5	<2	<2	9	.3	9	<2	5	.01	.006	<2	8	.06	23	.13	2	.29	.01	.16	3	1410
E 201307	3	10	28	8	8.8	8	6	22	2.59	132	<5	<2	<2	5	<.2	6	<2	4	.02	.003	<2	6	.02	18	.10	2	.26	.01	.16	3	1200
E 201308	4	10	17	11	6.0	15	10	41	4.76	137	<5	<2	<2	4	.7	6	<2	5	.01	.004	<2	8	.02	20	.12	4	.28	.01	.18	3	510
E 201309	7	34	81	90	17.0	17	12	89	7.11	156	<5	4	<2	22	.4	10	4	7	.12	.012	<2	9	.07	30	.12	5	.46	.03	.16	4	4040
RE E 201309	7	35	79	92	17.0	18	12	89	7.02	148	<5	3	<2	22	.8	10	<2	7	.12	.012	<2	10	.07	30	.12	2	.45	.03	.16	4	4090
E 201310	9	17	94	11	26.6	15	8	55	4.84	148	<5	4	<2	10	<.2	9	3	6	.17	.007	<2	9	.04	31	.19	4	.29	.01	.18	3	3910
E 201311	4	13	37	12	15.2	15	10	31	4.55	134	<5	<2	<2	5	.2	10	<2	4	.01	.004	<2	11	.02	25	.12	4	.26	.01	.18	3	1840
E 201312	7	34	130	169	14.3	18	13	52	6.09	121	<5	3	<2	11	.9	6	<2	6	.01	.011	2	6	.07	27	.15	3	.33	.01	.17	3	3750
E 201313	10	10	55	11	16.5	10	6	39	3.51	157	<5	2	<2	15	.2	10	<2	6	.04	.004	<2	8	.05	49	.19	4	.30	.04	.18	2	3720
E 201314	2	25	88	149	9.4	12	9	45	5.16	111	<5	4	<2	32	.7	5	<2	5	.04	.018	<2	10	.03	43	.14	<2	.31	.01	.19	1	4070
E 201315	2	24	31	83	6.0	25	15	46	6.91	114	<5	<2	<2	28	.4	3	<2	5	.09	.023	<2	6	.06	33	.11	<2	.37	.02	.17	1	1620
E 201316	6	11	12	10	4.1	24	15	25	6.52	108	<5	<2	<2	20	.5	3	2	5	.01	.019	<2	7	.02	36	.20	<2	.27	.01	.16	1	1310
E 201317	3	16	26	13	7.3	28	17	37	7.29	113	<5	<2	<2	9	.3	4	<2	4	.01	.005	<2	10	.03	19	.11	2	.26	.01	.14	1	1230
E 201318	2	10	31	8	9.6	26	14	22	7.79	119	<5	<2	<2	10	.4	7	<2	4	.01	.006	<2	6	.02	18	.08	<2	.27	.01	.15	1	1270
E 201319	5	19	50	7	13.8	23	16	39	9.43	147	<5	<2	<2	13	.7	13	<2	5	.01	.011	2	6	.04	19	.11	3	.29	.01	.14	1	1620
E 201320	4	18	43	13	10.6	23	14	63	7.63	141	<5	<2	<2	19	.6	12	<2	4	.03	.011	2	12	.06	18	.08	<2	.32	.01	.15	1	1570
E 201321	4	18	60	70	14.9	24	13	32	5.75	104	<5	<2	<2	27	<.2	21	<2	5	.04	.011	2	7	.03	23	.13	2	.31	.02	.13	3	1650
E 201322	4	10	77	40	14.9	24	15	29	6.92	142	<5	<2	<2	63	.3	16	<2	7	.07	.014	2	8	.03	39	.16	<2	.36	.03	.15	3	1760
E 201323	5	10	47	12	11.7	19	11	32	7.48	193	<5	2	<2	60	.3	11	<2	6	.02	.013	2	12	.03	34	.18	<2	.28	.04	.17	2	2530
E 201324	4	10	34	6	5.8	13	7	34	4.22	142	<5	<2	<2	37	<.2	4	<2	6	.01	.006	<2	10	.03	35	.15	2	.28	.04	.15	4	1440
E 201325	6	15	17	4	8.7	21	10	26	6.05	227	<5	<2	<2	4	.4	6	<2	4	.02	.005	<2	10	.03	22	.13	3	.25	.01	.15	3	2050
E 201326	15	18	61	15	6.2	18	13	58	6.84	56	<5	<2	<2	5	<.2	3	<2	5	.01	.002	<2	9	.07	32	.05	2	.33	.01	.17	1	1820
E 201327	3	14	3	13	.3	31	16	141	4.74	31	<5	<2	<2	23	<.2	2	4	5	.25	.017	2	7	.37	47	.16	4	.77	.01	.16	1	48
E 201328	52	70	10	87	1.6	21	9	469	7.11	29	12	<2	<2	382	.7	4	<2	79	.85	.038	3	47	.60	93	.31	2	2.36	.08	.10	<1	49
E 201329	49	53	24	104	1.0	34	18	167	6.60	51	9	<2	<2	100	<.2	3	2	21	.98	.033	<2	23	.16	61	.18	2	1.51	.11	.17	<1	69
E 201330	26	70	12	40	3.1	42	16	346	5.27	43	5	<2	<2	76	<.2	2	3	30	.73	.035	<2	37	.64	72	.14	3	1.74	.17	.17	2	140
E 201331	32	13	36	5	8.0	20	8	57	3.98	107	<5	<2	<2	32	.5	4	<2	7	.09	.010	<2	18	.07	66	.15	<2	.32	.04	.17	1	1480
E 201332	34	60	129	178	34.8	22	8	233	7.41	79	<5	9	<2	21	.7	15	2	26	.46	.008	<2	30	.24	56	.13	2	.64	.04	.11	2	8010
E 201333	9	30	69	106	10.9	10	2	468	1.17	12	<5	<2	<2	10	1.2	9	<2	21	2.67	.010	<2	14	.03	3	.08	4	.65	<.01	.01	3	260
E 201334	2	137	138	304	222.1	30	14	323	4.59	36	<5	7	<2	25	.5	3	<2	5	.34	.039	<2	9	.38	72	.02	<2	.81	.05	.17	1	6680
E 201335	4	21	20	11	26.2	22	11	47	4.26	46	<5	<2	<2	17	<.2	<2	<2	3	.01	.010	<2	9	.04	60	.01	<2	.32	.03	.18	1	1450
E 201336	8	2800	10264	40263	213.1	6	1	1614	.69	41	<5	40	<2	70	87.7	883	4	3	32.09	.007	2	2	.08	84	.01	2	.22	<.01	.01	<1	43800
E 201337	53	14	131	70	18.3	13	3	68	3.19	85	<5	<2	<2	43	<.2	8	4	7	.07	.009	2	10	.04	81	.19	3	.33	.05	.17	2	1340
STANDARD C/AU-R	17	58	37	123	6.7	68	29	1075	3.93	41	18	7	37	51	18.0	14	21	54	.51	.085	37	55	.92	190	.09	33	1.87	.06	.14	11	480

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.

THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS &gt; 1%, AG &gt; 30 PPM &amp; AU &gt; 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples Beginning 'RE' are duplicate samples.

DATE RECEIVED: SEP 27 1993

DATE REPORT MAILED: Sept 30/93

SIGNED BY: D.TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS



## GEOCHEMICAL ANALYSIS CERTIFICATE

Canasil Resources Inc. File # 93-2408

GRANITE Project



1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: Paul Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
C 24081	14	83	34	325	44.3	42	16	331	6.51	103	<5	<2	<2	51	.6	26	2	12	2.55	.044	<2	20	.32	12	.06	2	4.87	.59	.24	<1	280
C 24082	<1	20	14	101	9.8	27	11	8549	2.93	43	<5	<2	<2	37	.2	<2	2	23	13.57	.030	<2	20	.60	41	.05	<2	3.93	.18	.41	<1	470
C 24083	1	21	18	278	9.9	35	14	3613	3.79	46	<5	<2	<2	41	.7	<2	<2	26	6.11	.043	<2	26	.62	18	.07	<2	4.63	.29	.47	<1	400
C 24084	2	41	83	211	25.1	33	14	358	5.82	89	<5	<2	<2	17	<.2	12	<2	6	.45	.047	<2	18	.38	15	.08	2	1.01	.06	.34	<1	580
C 24085	3	127	400	753	83.8	37	13	321	6.22	136	<5	<2	<2	30	2.6	44	<2	6	.87	.042	<2	9	.22	13	.06	5	1.60	.17	.26	<1	850
C 24086	3	59	140	250	64.6	42	15	335	5.70	121	<5	<2	<2	41	.2	25	<2	7	1.24	.049	<2	10	.23	19	.07	5	2.17	.25	.26	<1	980
C 24087	2	39	100	269	29.5	40	12	2237	4.62	67	<5	9	<2	30	.3	3	<2	14	2.65	.041	<2	27	.41	21	.06	<2	2.49	.11	.27	<1	9040
C 24088	1	41	105	215	30.6	37	14	558	5.37	91	<5	6	<2	22	.6	3	<2	6	.97	.048	<2	9	.24	24	.05	2	1.34	.06	.22	<1	5140
C 24089	2	205	1112	2306	74.8	36	14	345	5.17	122	<5	5	<2	23	7.2	35	<2	5	.95	.049	<2	8	.23	16	.05	<2	1.42	.09	.21	<1	4080
C 24090	2	55	241	528	31.7	41	13	603	5.08	91	<5	2	<2	34	1.8	10	<2	7	1.80	.046	<2	19	.25	19	.06	<2	2.26	.18	.22	<1	2050
C 24091	1	27	16	126	12.9	34	13	1041	4.87	58	<5	2	<2	24	.3	<2	<2	10	1.63	.046	<2	12	.32	22	.06	<2	1.86	.07	.28	1	1870
RE C 24091	1	27	18	123	13.0	36	15	1059	5.04	62	<5	2	<2	24	<.2	<2	<2	10	1.68	.048	<2	12	.32	20	.06	3	1.87	.07	.28	1	1910
C 24092	2	26	26	133	13.6	43	14	1071	5.04	89	<5	3	<2	43	1.0	3	<2	11	2.21	.047	<2	11	.29	24	.07	2	2.12	.10	.22	<1	2080
C 24093	3	102	230	3582	220.7	37	12	225	7.46	101	<5	10	<2	65	12.6	22	<2	5	1.05	.040	<2	20	.14	7	.06	3	1.59	.12	.19	<1	9930
C 24096	2	24658	<2	1418	151.4	21	24	156	6.72	<2	<5	3	<2	7	34.7	6	<2	5	.63	.002	<2	10	.12	8	<.01	<2	.19	<.01	.02	<1	19500
C 24097	8	68	64	138	10.7	11	6	42	4.70	177	<5	3	<2	12	.3	9	<2	4	.04	.005	<2	8	.02	26	.12	3	.21	.01	.14	7	2280
C 24098	20	26	76	185	32.0	15	6	41	4.76	171	<5	8	<2	6	.8	14	2	4	.04	.002	<2	30	.02	29	.10	<2	.23	.01	.13	10	7030
C 24099	16	14	168	89	50.8	6	<1	28	.85	111	<5	8	<2	3	.4	23	<2	3	.01	.002	<2	9	.01	28	.10	<2	.19	.01	.13	3	7140
C 24100	7	9	62	7	17.5	8	3	31	2.42	132	<5	4	<2	3	.4	10	<2	3	.01	.002	<2	24	.01	31	.12	2	.20	.01	.15	4	3050
E 201301	6	13	105	15	20.6	10	3	26	2.09	92	<5	4	<2	10	.4	9	<2	3	.02	.003	<2	7	.01	24	.09	4	.16	.02	.14	2	3170
STANDARD C/AU-R	16	56	38	128	6.7	68	29	1031	3.97	43	17	6	36	51	16.8	14	17	55	.50	.085	35	55	.89	186	.09	32	1.88	.05	.13	11	510

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.

THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS &gt; 1%, AG &gt; 30 PPM &amp; AU &gt; 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: SEP 10 1993 DATE REPORT MAILED: Sept 15/93 SIGNED BY: D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS





## GEOCHEMICAL ANALYSIS CERTIFICATE

Canasil Resources Inc. File # 93-2403

GRANITE Project



1695 Marine Drive, North Vancouver BC V7P 1V1

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
C 24094	2	236	24	40	16.7	18	18	374	4.85	13	<5	<2	<2	122	1.0	<2	<2	47	2.19	.036	<2	19	.67	23	.08	<2	3.89	.41	.07	1	250
C 24095	1	138	5	58	1.9	14	20	617	5.19	9	<5	<2	<2	126	.6	<2	<2	26	2.36	.040	<2	8	.57	45	.06	<2	4.28	.43	.11	1	17
E 201302	8	13	84	60	15.0	8	3	25	2.68	124	<5	5	<2	7	.2	10	<2	4	.02	.002	<2	10	.01	36	.12	<2	.22	.01	.18	4	4480
RE E 201302	8	13	86	62	16.0	10	2	28	2.73	131	<5	6	<2	7	<.2	11	<2	4	.02	.003	<2	11	.01	37	.12	<2	.23	.01	.19	4	4710

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.  
THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: SEP 10 1993

DATE REPORT MAILED:

Sept 14/93.

SIGNED BY.....D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS



## GEOCHEMICAL ANALYSIS CERTIFICATE

Canasil Resources Inc. File # 93-2132

GRANITE Project



1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: P.J. Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
C 24068	18	41	73	87	7.9	38	15	475	3.93	111	8	<2	<2	66	<2	10	<2	12	.33	.047	<2	15	.67	126	.12	3	1.22	.07	.23	<1	510
C 24069	24	206	618	2010	431.7	32	14	1380	10.90	318	8	<2	<2	27	2.1	107	<2	20	2.40	.027	<2	16	.27	89	.06	<2	3.44	.42	.23	<1	3070
C 24070	5	61	35	108	5.0	37	20	1163	4.73	66	5	<2	<2	115	.2	5	2	79	2.58	.033	<2	49	1.37	61	.12	<2	5.42	.65	.19	1	76
RE C 24070	4	59	31	102	4.8	34	19	1115	4.50	62	<5	<2	<2	112	<2	4	<2	75	2.47	.031	<2	46	1.30	59	.12	<2	5.15	.62	.19	<1	71
C 24071	34	90	237	703	177.5	34	14	367	17.48	291	7	<2	2	30	<2	51	<2	12	.85	.029	<2	14	.30	40	.08	<2	2.99	.40	.24	<1	2260
C 24072	2	40	19	188	14.6	11	16	633	4.99	17	<5	<2	<2	79	<2	2	<2	57	1.58	.032	<2	12	1.07	67	.13	<2	2.72	.31	.15	2	110
C 24073	20	102	6	78	1.1	12	18	698	5.89	4	<5	<2	<2	135	<2	<2	5	71	1.36	.031	<2	17	1.31	43	.12	<2	2.99	.36	.05	<1	14
C 24074	3	122	5	42	1.2	5	11	472	3.13	<2	<5	<2	<2	58	<2	<2	<2	51	1.41	.067	<2	6	.87	176	.15	<2	1.92	.13	.04	<1	18
C 24075	1	54	2	49	.3	14	15	471	3.23	<2	<5	<2	<2	100	<2	<2	<2	61	1.26	.044	<2	19	1.07	114	.15	2	2.57	.24	.07	<1	4
C 24076	5	23	41	8	17.2	16	11	25	3.88	120	<5	<2	<2	7	<2	11	<2	5	.04	.004	<2	6	.02	20	.08	4	.28	.01	.15	3	1810
C 24077	79	26	251	64	69.8	26	13	62	8.49	143	<5	6	2	15	<2	34	6	6	.06	.007	<2	7	.05	30	.07	2	.43	.02	.18	8	5560
C 24078	38	104	12	155	6.4	30	10	792	1.81	21	8	<2	<2	39	.6	3	<2	19	9.04	.020	2	7	.02	6	.09	2	.49	<.01	.01	2	1210
C 24079	5	134	11	47	5.8	76	27	533	4.43	30	<5	<2	<2	100	<2	<2	<2	42	1.95	.023	<2	57	1.12	63	.12	<2	3.26	.30	.02	1	550
C 24080	<1	1727	2	91	14.4	50	93	1114	20.14	<2	11	<2	2	15	<2	<2	<2	20	2.48	.032	<2	9	.35	4	.03	<2	.67	.01	<.01	<1	520
STANDARD C/AU-R	17	58	38	126	6.8	67	31	1037	3.96	39	18	6	36	52	17.5	14	22	54	.51	.086	37	56	.91	183	.08	33	1.88	.06	.14	10	460

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.

THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS &gt; 1%, AG &gt; 30 PPM &amp; AU &gt; 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: AUG 25 1993

DATE REPORT MAILED:

Aug 30/93

SIGNED BY: .....D.TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS



## GEOCHEMICAL ANALYSIS CERTIFICATE



Canasil Resources Inc. File # 93-1877

GRANITE Project

1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: P.J. Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
C 24059	2	119	3	109	.8	11	18	1022	4.71	22	<5	<2	<2	84	1.9	<2	<2	41	.98	.036	<2	11	.92	61	.14	4	1.72	.10	.07	1	17
RE C 24059	2	118	4	110	.8	9	17	1014	4.62	15	<5	<2	<2	83	2.2	<2	<2	41	.98	.037	<2	11	.91	59	.14	<2	1.71	.10	.07	2	17
C 24060	3	104	4	126	.9	13	17	1086	4.53	10	<5	<2	<2	102	1.2	<2	<2	39	2.42	.035	<2	13	.75	73	.14	2	2.40	.19	.09	2	20
C 24061	3	108	<2	37	.2	8	13	849	2.97	14	<5	<2	<2	71	.2	<2	<2	53	1.65	.056	2	11	1.34	33	.15	<2	2.48	.09	.06	3	6
C 24062	1	151	<2	25	.2	5	7	601	3.15	13	<5	<2	<2	52	.6	<2	<2	49	.85	.075	3	5	1.15	48	.14	<2	1.42	.05	.10	3	4
C 24063	10	264	2	39	.3	28	14	623	4.94	18	<5	<2	<2	48	.5	<2	<2	64	.85	.059	3	38	1.18	36	.19	<2	1.50	.08	.23	2	4
C 24064	7	105	13	228	.9	37	14	1091	4.20	53	<5	<2	<2	60	1.3	<2	<2	49	.99	.044	2	53	1.72	53	.17	<2	2.20	.06	.04	2	21
C 24065	3	19	19	21	4.8	19	12	36	4.78	111	<5	<2	<2	12	<.2	2	3	4	.03	.005	<2	7	.03	24	.09	<2	.26	.01	.13	3	1060
C 24066	6	50	7	30	.6	52	22	181	5.39	29	<5	<2	<2	28	.4	<2	4	6	.27	.057	<2	7	.41	57	.07	4	1.00	.03	.19	2	94
C 24067	28	109	16	105	3.7	42	20	397	7.09	43	7	<2	<2	103	.5	2	<2	31	1.16	.040	2	26	.37	35	.16	<2	1.90	.16	.15	2	270
STANDARD C/AU-R	17	58	34	124	6.9	64	29	1050	3.96	42	20	7	35	55	17.9	14	20	53	.51	.086	38	54	.91	194	.09	33	1.88	.06	.14	10	460

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.

THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS &gt; 1%, AG &gt; 30 PPM &amp; AU &gt; 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: AUG 9 1993

DATE REPORT MAILED:

Aug 11/93.

SIGNED BY: *C. Leong* .D.TOYE, C.LEONG, J.WANG; CERTIFIED B.C. ASSAYERS



## GEOCHEMICAL ANALYSIS CERTIFICATE

GRANITE Project



Canasil Resources Inc. File # 93-1753  
 1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: P.J. Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
C 24051	9	34	9	66	.3	32	15	881	4.83	14	<5	<2	<2	44	<.2	<2	<2	74	3.31	.037	3	56	2.03	43	.03	3	2.64	.08	.11	<1	13
C 24052	1	223	640	1829	546.7	38	14	373	5.09	134	<5	21	<2	52	5.6	90	<2	11	1.79	.042	<2	13	.26	118	.10	<2	2.89	.28	.34	<1	19400
C 24053	28	95	29	102	13.8	47	16	1701	4.82	107	8	<2	<2	112	.5	5	2	47	4.42	.035	4	38	.85	75	.15	<2	6.05	.77	.71	1	640
C 24054	7	305	696	2231	386.1	38	17	813	4.26	223	<5	4	<2	37	4.9	149	2	15	2.89	.041	<2	16	.30	284	.09	4	3.96	.49	.44	<1	3640
C 24055	18	59	77	137	49.5	54	21	1042	5.85	209	7	<2	<2	141	<.2	23	<2	35	3.92	.054	2	37	.49	56	.12	<2	6.10	.83	.37	<1	520
C 24056	2	53	7	81	2.8	29	16	1175	4.46	37	<5	<2	<2	138	<.2	2	2	78	2.94	.039	<2	64	1.65	116	.13	3	5.60	.63	.06	1	43
C 24057	1	32	30	180	16.4	35	15	449	4.79	89	<5	3	<2	32	.3	3	<2	11	1.16	.052	<2	11	.28	209	.07	5	2.08	.11	.40	<1	2140
RE C 24057	1	31	27	181	15.7	33	15	435	4.63	83	<5	3	<2	31	.6	3	<2	11	1.13	.049	<2	11	.27	203	.07	3	2.03	.10	.39	1	2060
C 24058	7	54	3	72	1.4	12	15	588	4.61	14	<5	<2	<2	93	.5	<2	<2	73	1.45	.035	<2	15	1.25	79	.15	3	2.99	.34	.15	1	30
STANDARD C/AU-R	18	56	38	122	6.6	68	28	989	3.96	41	18	7	37	54	17.0	14	20	56	.50	.086	37	55	.92	183	.09	35	1.88	.06	.14	11	460

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.  
 THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.  
 ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS > 1%, AG > 30 PPM & AU > 1000 PPB  
 - SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: AUG 3 1993

DATE REPORT MAILED:

Aug 5/93.

SIGNED BY: *C. Leong* D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS



## GEOCHEMICAL ANALYSIS CERTIFICATE

GRANITE Project



Canasil Resources Inc. File # 93-0437

1695 Marine Drive, North Vancouver BC V7P 1V1 Submitted by: Paul Weishaupt

SAMPLE#	Mo ppm	Cu ppm	Pb ppm	Zn ppm	Ag ppm	Ni ppm	Co ppm	Mn ppm	Fe %	As ppm	U ppm	Au ppm	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	V ppm	Ca %	P %	La ppm	Cr ppm	Mg %	Ba ppm	Ti %	B ppm	Al %	Na %	K %	W ppm	Au* ppb
3518	12	289	729	818	210.8	35	22	1827	6.44	333	<5	3	<2	61	2.2	119	<2	27	4.51	.032	2	27	.43	13	.09	<2	4.49	.60	.31	2	3260
3519	3	138	1350	3726	63.1	45	21	445	5.37	103	<5	3	<2	27	12.9	18	<2	11	1.50	.046	2	24	.27	14	.08	<2	2.17	.17	.25	<1	3670
3520	2	62	33	118	4.6	33	25	590	6.10	31	<5	<2	<2	106	.5	3	<2	71	1.96	.038	<2	64	1.11	30	.16	<2	3.70	.42	.05	2	130
RE 3520	2	63	32	112	3.6	32	25	555	6.06	33	<5	<2	<2	106	.2	3	<2	71	1.94	.039	<2	65	1.11	35	.16	<2	3.55	.42	.05	2	150
STANDARD C/AU-R	17	62	38	130	6.9	68	32	1031	3.96	42	18	7	36	54	18.6	13	19	55	.50	.087	39	60	.90	187	.09	35	1.88	.07	.14	14	460

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.

THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL.

ASSAY RECOMMENDED FOR ROCK AND CORE SAMPLES IF CU PB ZN AS &gt; 1%, AG &gt; 30 PPM &amp; AU &gt; 1000 PPB

- SAMPLE TYPE: ROCK AU\* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE. Samples beginning 'RE' are duplicate samples.

DATE RECEIVED: MAR 10 1993

DATE REPORT MAILED:

Mar 16/93

SIGNED BY:

D. TOYE, C. LEONG, J. WANG; CERTIFIED B.C. ASSAYERS

APPENDIX 3

PROJECT COST

July 5, 1993 to October 2, 1993.

1.	<u>Wages</u>	H. Stirnimann 48 days @ \$165	\$7,920	
		P. Weishaupt 48 days @ \$180	8,640	\$16,560
2.	<u>Analysis</u>	87 Rock samples @ \$13.35		1,118
3.	<u>Campcost</u>	96 man days @ \$30		2,880
4.	<u>Freight</u>			120
5.	<u>Truck expense</u>	1.5 months @ \$950		1,425
6.	<u>Drilling and Blasting</u>	Cobra drill, drill steel, powder and fuses		450
				<hr/>
		Total field cost of project		\$ 22,553

APPENDIX 4

STATEMENT OF QUALIFICATIONS

NAME: P.J. WEISHAAPT

EDUCATION: Graduated Institute of Technology Agriculture  
Flawil, Switzerland.

AFFILIATIONS: Member Canadian Institute of Mining  
The Geological Society  
Member Geological Association of Canada

EXPERIENCE:

1960 - 1967	Bralorne-Pioneer Mines Prospector, Geologist's assistant, Underground mining and surveying.
1968 - 1970	Can-Fer Mines Ltd. Geologist.
1970 - 1973	Bralorne Resources Ltd. Exploration Manager.
1973 - 1975	Westfour Contracting Ltd. Manager, Coal Division.
1975 - 1977	Dolmage, Mason & Stewart Consulting Project Manager.
1978 - 1981	McIntyre Coal Mine Environmental Consultant
1981 - to present	Canmine Development Company Inc. & Canasil Resources Inc. President.

CANASIL RESOURCES

INDEX MAP  
of  
BRITISH COLUMBIA

☼ GRANITE M.C.

160 120 80 40 0 80 160 240  
Kilometers

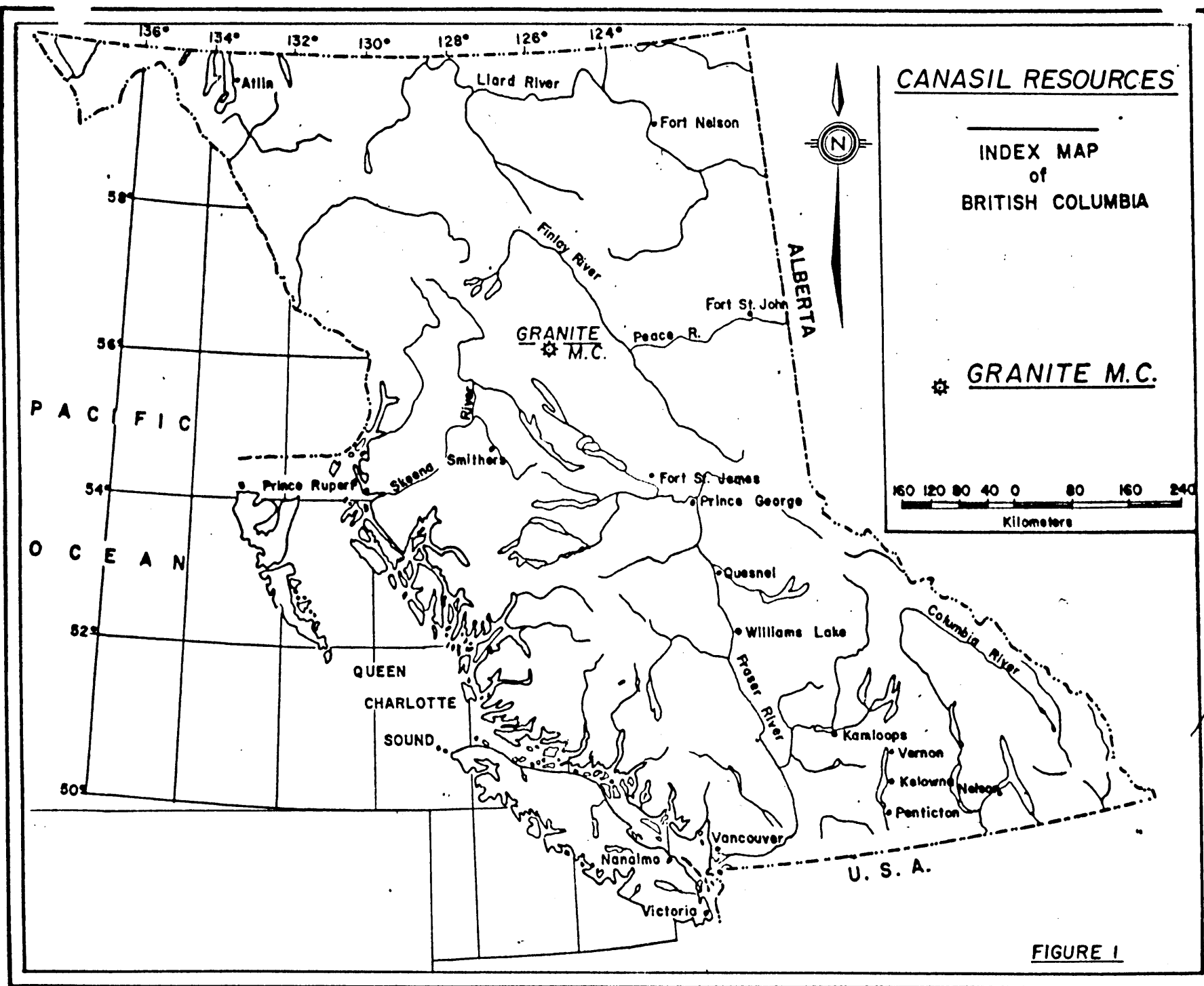
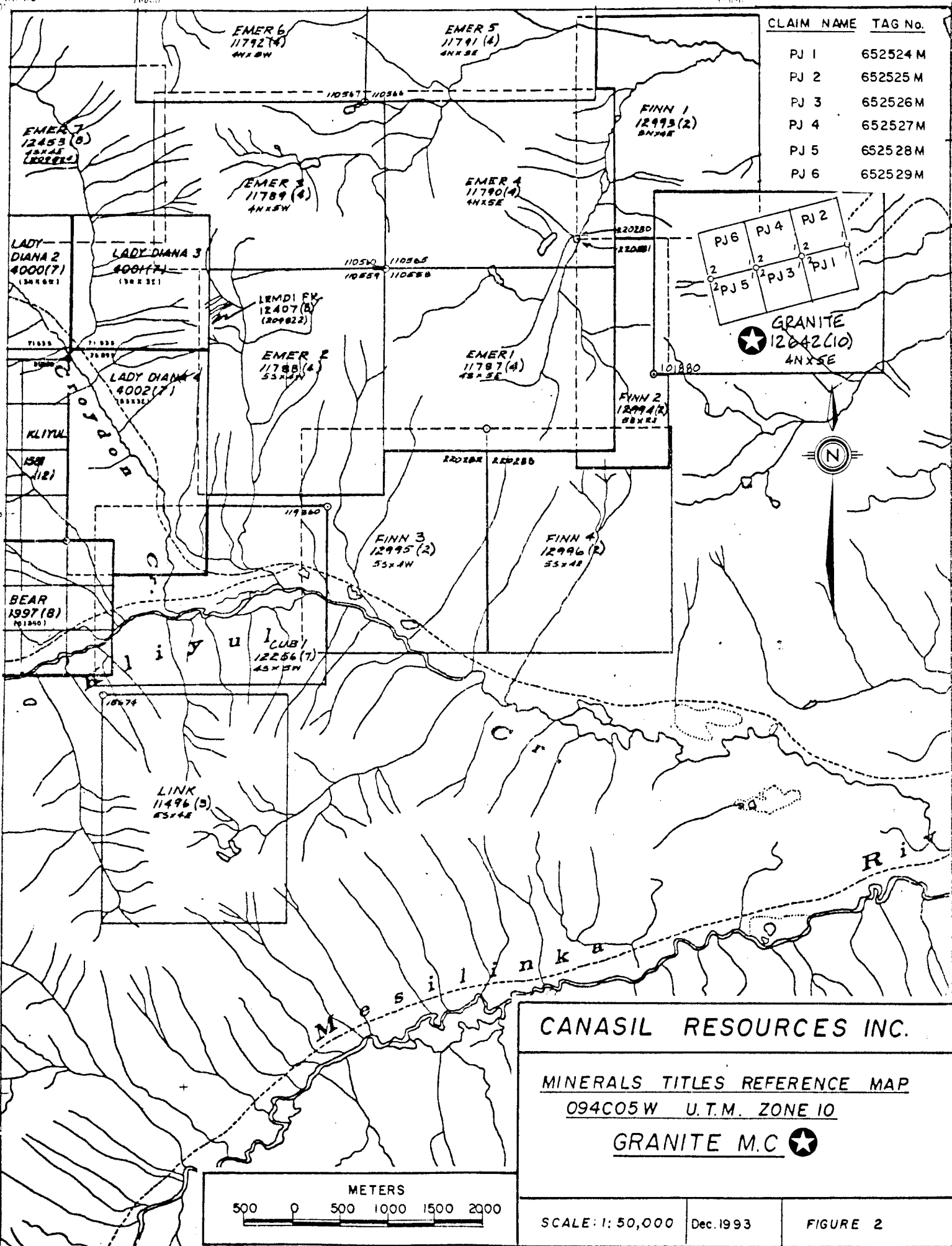


FIGURE 1



1059950

CLAIM NAME	TAG No.
PJ 1	652524 M
PJ 2	652525 M
PJ 3	652526 M
PJ 4	652527 M
PJ 5	652528 M
PJ 6	652529 M



CANASIL RESOURCES INC.

MINERALS TITLES REFERENCE MAP

094C05W U.T.M. ZONE 10

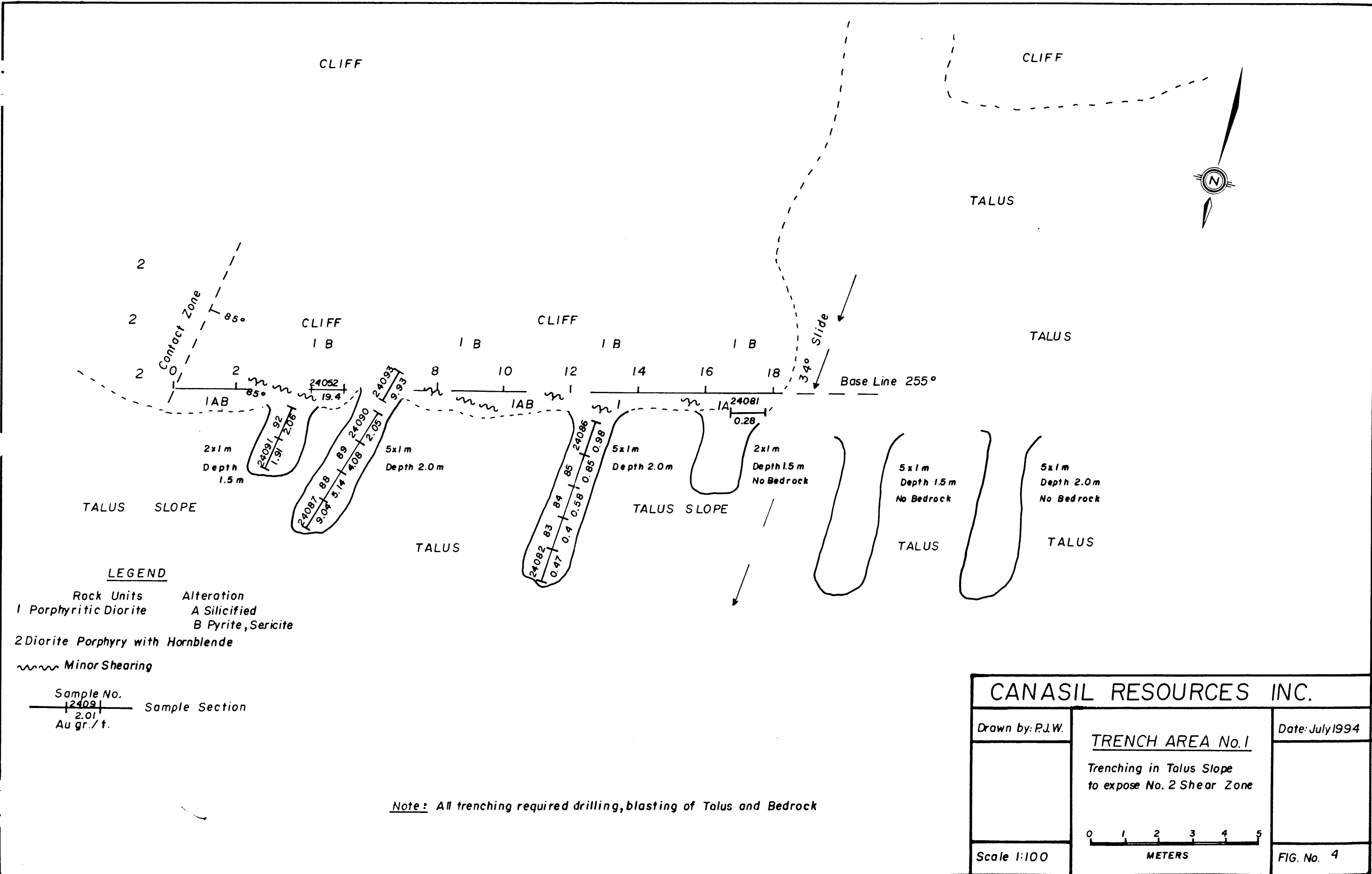
GRANITE M.C. ★

SCALE: 1: 50,000

Dec. 1993

FIGURE 2

6253632



CLIFF

CLIFF

TALUS



2

2

2

CLIFF  
I B

CLIFF  
I B

TALUS

Contact Zone  
85°

Slide  
34°

Base Line 255°

2x1m  
Depth 1.5m

5x1m  
Depth 2.0m

5x1m  
Depth 2.0m

2x1m  
Depth 1.5m  
No Bedrock

5x1m  
Depth 1.5m  
No Bedrock

5x1m  
Depth 2.0m  
No Bedrock

TALUS SLOPE

TALUS SLOPE

TALUS

TALUS

**LEGEND**

- Rock Units
- 1 Porphyritic Diorite
- 2 Diorite Porphyry with Hornblende
- Alteration
- A Silicified
- B Pyrite, Sericite
- Minor Shearing

Sample No. 2409 | Sample Section  
2.01 Au gr./t.

Note: All trenching required drilling, blasting of Talus and Bedrock

**CANASIL RESOURCES INC.**

Drawn by: P.J.W.

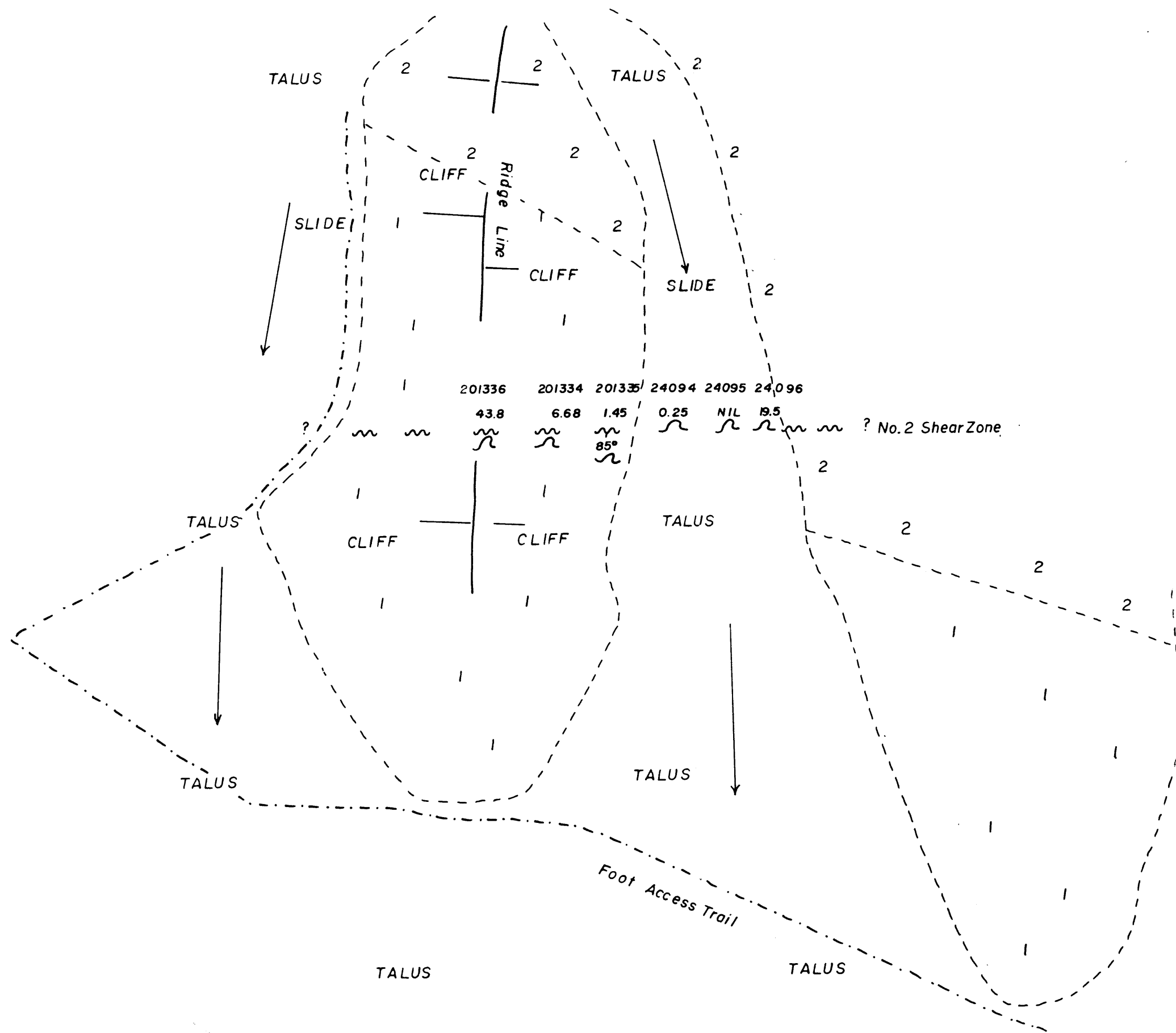
Date: July 1994

TRENCH AREA No. 1  
Trenching in Talus Slope  
to expose No. 2 Shear Zone



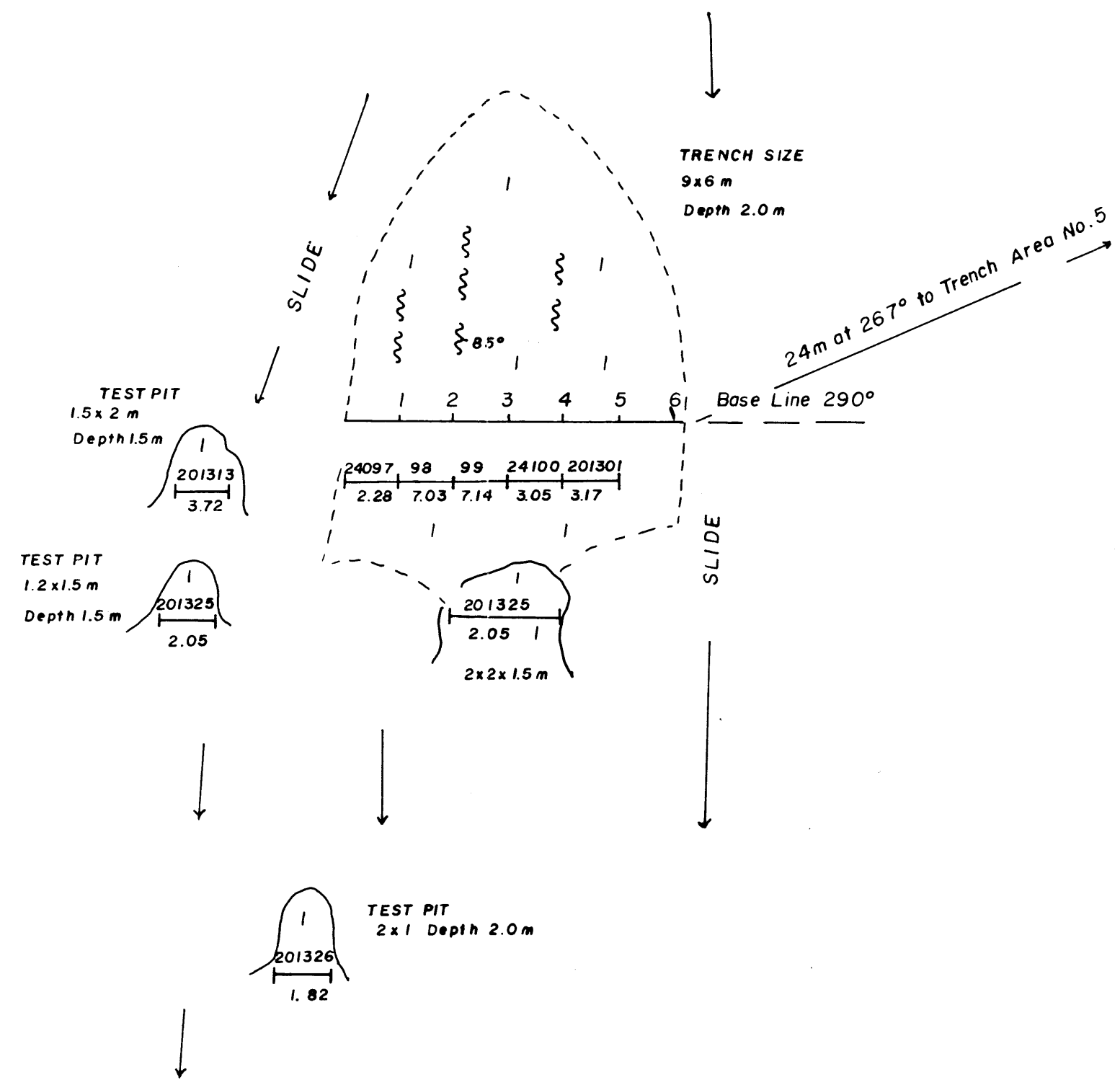
Scale 1:100

FIG. No. 4



**LEGEND**  
 1 Porphyritic Diorite silicified, pyritized  
 2 Tuff, Argillite and Limestone  
 24096 Sample No.  
 19.5 Au gr/t  
 Trenches drilled and blasted

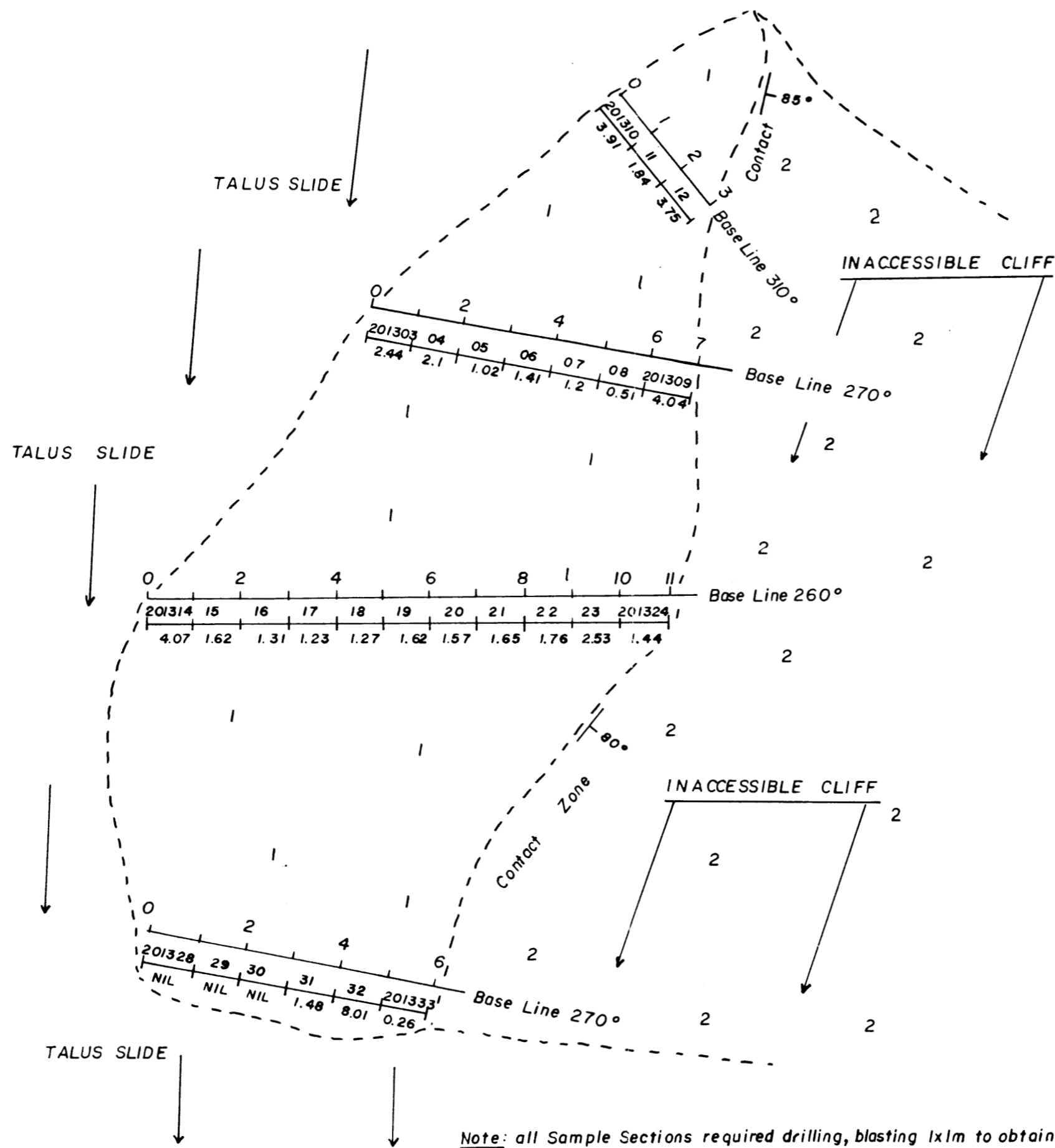
<b>CANASIL RESOURCES INC.</b>		
Drawn by: P.J.W.	<b>TRENCH AREA No.2,3</b>	Date: July 1994
	Trenching to expose No. 2 Shear Zone	
Scale 1:500	METERS	FIG.No. 5



**LEGEND**

- | Porphyritic Diorite silicified, bleached and pyritized
- ~ Minor Shearing
- Sample No. 20130
- Sample Section
- Au gr./t. 3.17

<b>CANASIL RESOURCES INC.</b>		
Drawn by: P.J.W.	<b>TRENCH AREA No. 4</b>	Date: July 1994
	Trenching to expose No. 3 Shear Zone	
	<p>METERS</p>	
Scale 1:100		FIG. No. 6

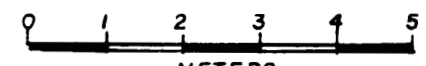


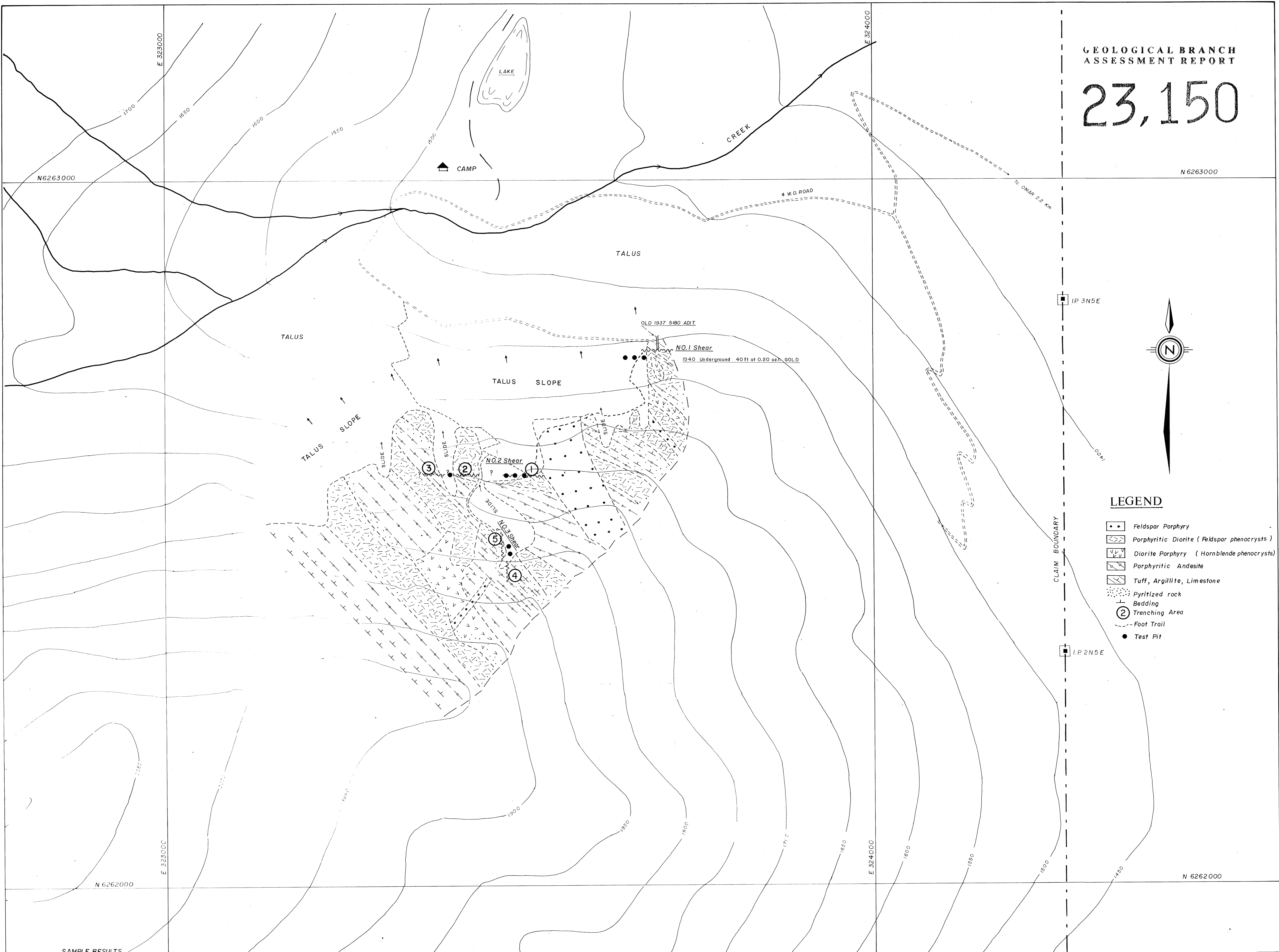
**LEGEND**

- 1 Porphyritic Diorite, silicified, bleached, pyritized
- 2 Tuff, Argillite and Limestone

Sample No.  
 201310  
 3.91  
 Au gr./t.  
 Sample Section

Note: all Sample Sections required drilling, blasting 1x1m to obtain Samples

<b>CANASIL RESOURCES INC.</b>		
Drawn by: P.J.W.	<b>TRENCH AREA No. 5</b>	Date: July 1994
	Trenching to expose No. 3 Shear Zone	
Scale 1:100	 METERS	FIG. No. 7

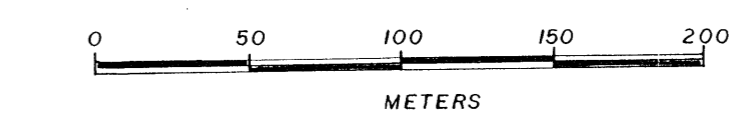


- LEGEND**
- Feldspar Porphyry
  - ▨ Porphyritic Diorite (Feldspar phenocrysts)
  - ▧ Diorite Porphyry (Hornblende phenocrysts)
  - ▩ Porphyritic Andesite
  - ▤ Tuff, Argillite, Limestone
  - ⋯ Pyritized rock
  - ⊥ Bedding
  - ② Trenching Area
  - Foot Trail
  - Test Pit

**SAMPLE RESULTS**

NO. 2 SHEAR					NO. 3 SHEAR				
Trench Area	SAMPLE NO.	WIDTH in meters	Au gr/t	Ag gr/t	Trench Area	SAMPLE NO.	WIDTH in meters	Au gr/t	Ag gr/t
Trench Area 1	24087 - 24090	4.0	5.5	41.6	Trench Area 4	24097 - 24100	5.0	5.5	29.5
	24091 - 24093	3.0	4.64	82.4		Trench Area 5	201301 - 201302		
	24052	1.2	19.4	546.7	201314 - 201324		11.0	1.8	9.8
Trench Area 2	201334	1.2	6.68	222.1	201303 - 201309		7.0	1.8	10.6
	201335	2.0	1.45	26.1	201310 - 201312	3.0	3.1	18.7	
	201336	0.5	43.8	213.1	201331 - 201332	2.0	4.7	22.8	
Trench Area 3	24096	0.7	19.5	151.4					

N.T.S. 94C/5W



Elevations in meters above Mean Sea Level  
Contour Interval - - - - - 50 Meters  
Grid System - - - - - U.T.M. Grid

**CANASIL RESOURCES INC.**

GRANITE MC.  
(GRANITE BASIN)

SAMPLE LOCATION and GEOLOGY

1993

DATE: DECEMBER 1993

Figure 3

SCALE: 1: 2500

DRAWN BY: P.J.W.

DRAW NO. G 93-01