



3/88

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TRENCHING, GEOLOGICAL MAPPING & SAMPLING,
AND DIAMOND DRILLING PROGRAMMES
ON THE
SADIM PROPERTY
SADIM 1 - 6 CLAIMS

Missezula Mountain Area
Similkameen Mining Division, B.C.

NTS Ref. 92H/10E

Latitude: 49°~~44'~~ 43'

Longitude: 120°~~30'~~ 32.5'

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For
Owner/Operator: LARAMIDE RESOURCES LTD.

By
I.M. WATSON & ASSOCIATES LTD.

I.M. Watson, P.Eng.
Vancouver, B.C.

GEOLOGICAL BRANCH
ASSESSMENT REPORT

15,969

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INTRODUCTION

The SADIM 1 - 5 claims are situated in the Missezula Mountain area of southwestern B.C. The claims are underlain by rocks of the Nicola Belt in a geological setting essentially similar to that hosting the porphyry copper-gold deposits of the Quesnel Trough in the Quesnel-Cariboo area.

Geological and geochemical reconnaissance surveys of the SADIM 1 - 4 claims during the summer of 1985 revealed gold and silver bearing quartz veins within a northerly trending zone of altered tuffs close to the common boundary between the SADIM 3 and 4 claims (Watson, 1985). Preliminary sampling of the veins and host rocks gave encouraging results; follow-up trenching programmes during September/October and December 1986 led to a preliminary six-hole diamond drilling programme during January/February 1987.

This report summarises the results of the 1986 trenching and 1987 drilling programmes.

LOCATION, ACCESS & PHYSIOGRAPHY (Figures 1 and 2)

The SADIM claims are situated four kilometres east of Highway 5, 30 kms. north of Princeton and 45 kms. south of Merritt, within the Similkameen Mining Division. The centre of the property is at $49^{\circ}44'40''\text{N}$, $120^{\circ}30'40''\text{W}$. The NTS reference is 92H/10E.

Access to the property from Highway 5 is by the Dillard-Ketchan Creek main logging roads which branch east from the highway about 12 kms. south to the village of Aspen Grove. The Ketchan Creek road traverses the SADIM 1 and 3 claims in a southeasterly direction. Distance from Highway 5 to the property is approximately 16 kms.

An alternate access route is by gravel logging road from Highway 5 at a point 2.5 kms. north of Allison Lake. This road climbs east for 5 kms. to join the Ketchan Creek road at the northwestern corner of the SADIM 1 claim.



LARAMIDE RESOURCES LTD.

Figure 1: Index Map

L. M. Watson & Associates Ltd.

Within the property boundaries, logging and 'mining' roads, and the B.C. Telephone microwave tower road, provide good access to all parts of the claim group. The B.C. Hydro power line crosses the centre of the SADIM 1 and 3 claims.

The property occupies the summit area of the broad, north trending ridge separating the deep fault valleys of Summers Creek to the east and Allison Creek to the west. Elevations on the property range from 1615 metres at the summit of Microwave Hill, on the common boundary between SADIM 1 and 2, to 1200 metres at the headwaters of Allison Creek, in the northwestern corner of the SADIM 1 claim. The topography is typical of this part of the Thompson Plateau, reflecting the effects of a predominantly northerly structural trend, accentuated by glaciation; heavily forested, relatively gentle upland slopes are cut by deep, steep-sided, north trending valleys. Bedrock exposure varies and is largely a function of glacial action; generally outcrop is abundant on ridges and along the upper slopes of steep valleys but lower slopes and valley bottoms bear a thick mantle of glacial overburden.

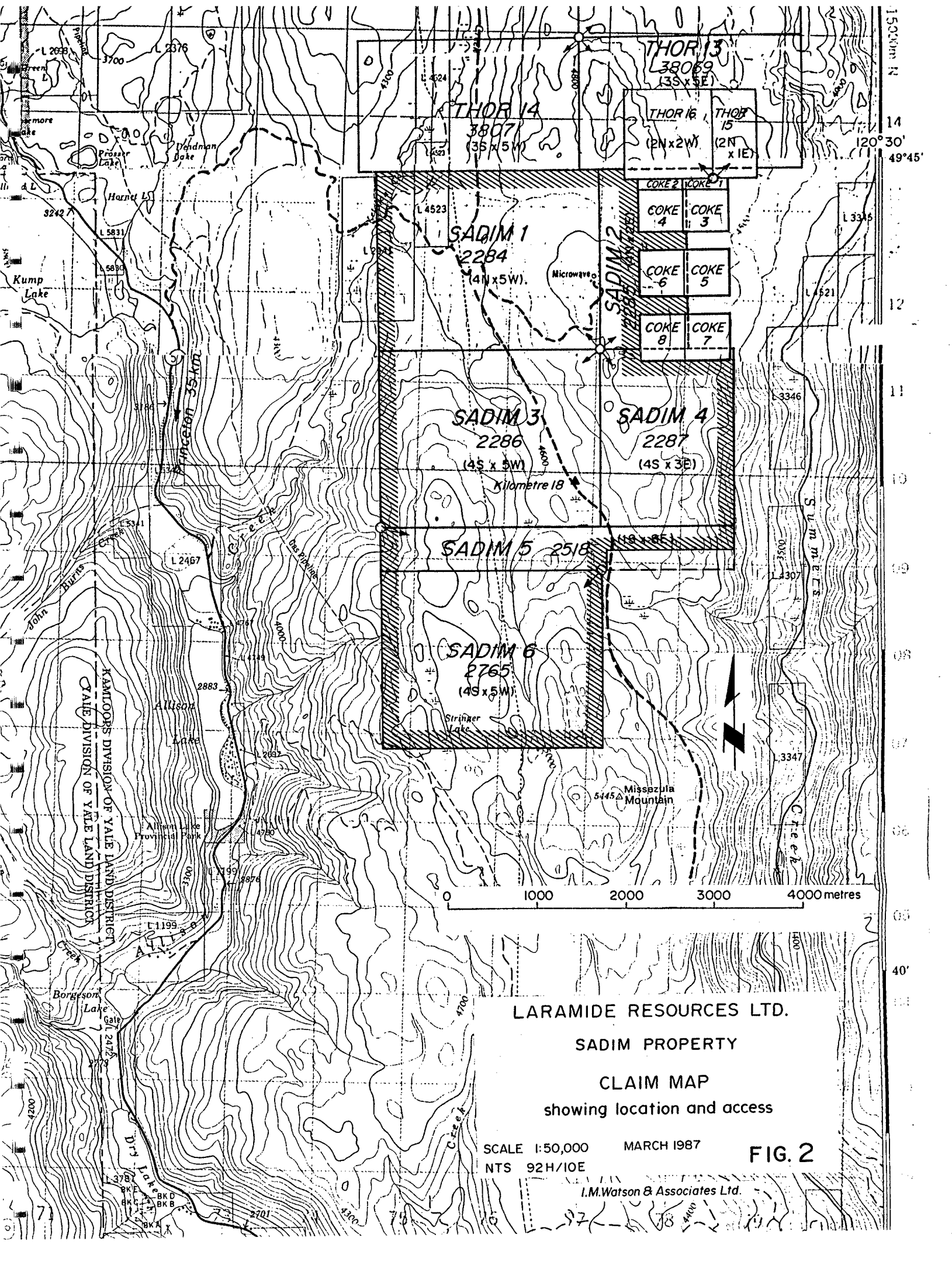
Away from the main north-south river valleys, drainage is weakly developed and consists of ill-defined water courses and seepages.

Vegetation is dense on shaded and northerly slopes, but is more open on south facing hillsides; mixed conifers, alder and poplar predominate. Logging operations are currently active immediately south of the SADIM claims.

CLAIMS (Figure 2)

The SADIM property consists of six mineral claims containing a total of 88 units, as follows:

<u>Claim Name</u>	<u>No. of Units</u>	<u>Record No.</u>	<u>Recording Date</u>
SADIM 1	20	2284	10 October 1984
SADIM 2	8	2285	10 October 1984
SADIM 3	20	2286	10 October 1984
SADIM 4	12	2287	10 October 1984
SADIM 5	8	2518	30 December 1985
SADIM 6	20	2765	8 December 1986



LARAMIDE RESOURCES LTD.
 SADIM PROPERTY
 CLAIM MAP
 showing location and access

SCALE 1:50,000 MARCH 1987
 NTS 92H/10E

FIG. 2

I.M. Watson & Associates Ltd.

The SADIM 1-4 claims were staked by and on behalf of I.M. Watson on the 17th and 18th September 1984. Ownership was transferred to Laramide Resources Ltd. by bill of sale dated 12 November, 1985. The SADIM 5 and 6 claims were staked for Laramide on 5th December, 1985 and 30th November, 1986, respectively.

HISTORY

The earliest record of work in the SADIM claim area dates back to the early 1960's - the beginning of the porphyry copper exploration boom which persisted until the early 1980's. Most of the work recorded within the present SADIM claim area was concentrated in the northeastern and eastern part of the claim group, over the SADIM 2 and 4 claims.

The following is a summary of past activity in the property area.

- 1962 The 40 claim KR group was staked by Plateau Metals Ltd. Work consisted of a magnetometer survey, bulldozer trenching, and an undisclosed amount of diamond drilling. The claims occupied the area presently covered by the SADIM 2 claim, and by the northern part of the SADIM 4 claim.
- 1966 Adera Mining Ltd. optioned the KR claims and carried out soil sampling and magnetometer surveys followed by diamond drilling. The claims were allowed to lapse.
- 1970 Amax Explorations Inc. staked the RUM claims; the southern half of the property lay within the area now covered by the SADIM 2 and 4 claims. Work done by Amax consisted of geological mapping, soil sampling, and magnetometer and I.P. surveys, followed by a nine-hole, 1879-foot percussion drilling programme.
- 1972 Kalco Valley Mines Ltd. optioned the RUM claims, then relinquished the property after a programme of mapping and trench sampling.

- 1973-74 Bronson Mines Ltd. staked the CINDY claims, covering ground now lying within the SADIM 1 claim. Mapping and prospecting programmes were carried out.
- 1974 Ruskin Developments Ltd. acquired the RUM claims, and completed geological mapping and soil sampling surveys before allowing the ground to lapse.
- 1979-81 Cominco Ltd. staked 55 claims, (RUM 1-55), coincident with the main area of interest covered by the original RUM claims staked by Amax. Cominco refurbished and renumbered the old Amax grid and used it for control of geological, soil and rock geochemical, and magnetometer surveys. Since then Cominco has allowed the claims to lapse.
- 1984-85 The SADIM 1-4 claims were staked by I.M. Watson and subsequently transferred to Laramide Resources in 1985. In the fall of 1985 Laramide carried out reconnaissance mapping, prospecting and geochemical soil sampling. Encouraging soil and rock geochemical results in the southern part of the SADIM 3 and 4 claims led to detailed sampling and mapping of the anomalous area. Gold and silver bearing quartz veins were found in rusty altered tuffs, over a strike distance of at least 300 metres. Preliminary chip sampling of the mineralised vein material yielded assays of up to 4,120 ppb Au, and a selected grab sample contained 0.20 oz/ton Au.

SUMMARY OF WORK SEPTEMBER 1986 - FEBRUARY 1987

The area of gold mineralisation discovered in 1985 is situated 200 metres east of the Ketchan Creek logging road at kilometre 18. Recent logging (1983) has bared a triangular area, about 1 kilometre long and 500 metres wide at the north end of the clearing. The topographic grain is northerly. A shallow, south draining swampy depression bisects the cleared area. To the east the ground rises moderately from the edge of the clearing to the summit of the broad ridge separating Allison and Summer Creeks. The mineralised veins and host tuff are intermittently exposed along the

eastern side of the swamp over a distance of 300 metres. Other, smaller exposures of mineralised veins have been found up to 700 metres to the north, but the 1986-87 trenching, sampling and drilling programmes were confined to the larger southern area of interest.

(a) Trenching/Sampling Programme

Phase I - Sept. 23 to Oct. 2, 1986 (Figures 4, 6a, and 7a)

Ten trenches were completed using a JWSB H90 Excavator rented from Douglas Lake Ranch Co. Trenches were laid out east-west across the general strike of lithology in the showing area. Seven trenches (Nos. 1-7) were spaced at 25 metre intervals along a 160 metre strike length. Host rock exposures a further 100 metres to the north and south were investigated by trenches #9 and #10 (north) and #8 (south). The total length trenched during Phase I was 320 metres. Average depth of the trenches was 1.5 metres, and overburden cover ranged from zero to 2.0 metres.

All trenches were sampled and mapped; continuous chip samples were taken at 1 metre intervals. Additional chip and channel samples were taken to test individual veins or zones of specific interest. A total of 186 samples was collected, and shipped to Acme Analytical Laboratories Ltd. in Vancouver, to be analysed for Au, Ag, Pb and Cu. Gold was determined by atomic absorption (AA), and silver, lead and copper by the inductively coupled argon plasma method (ICP).

Phase II - December 10 to 17, 1986 (Figures 4, 6a-c, 7a-c)

The Phase II trenching programme was undertaken to test the extent and tenor of the gold bearing quartz vein stockwork revealed by Phase I. Trenches #'s 2 - 7 were extended to the east and the ground to the north and northeast of the anomalous zone was tested by new trenches 1A and

11; trenches, 12, 13 and 14 explored the area west of the swamp. Total length of trenches completed, mapped and sampled during Phase II was 455 metres. Average depth was approximately 1.5 metres. 175 chip samples were sent to Acme Laboratories for analysis.

b) Diamond Drilling Programme - January 24th - February 11th, 1987
(Figs. 4, 5 and 6a)

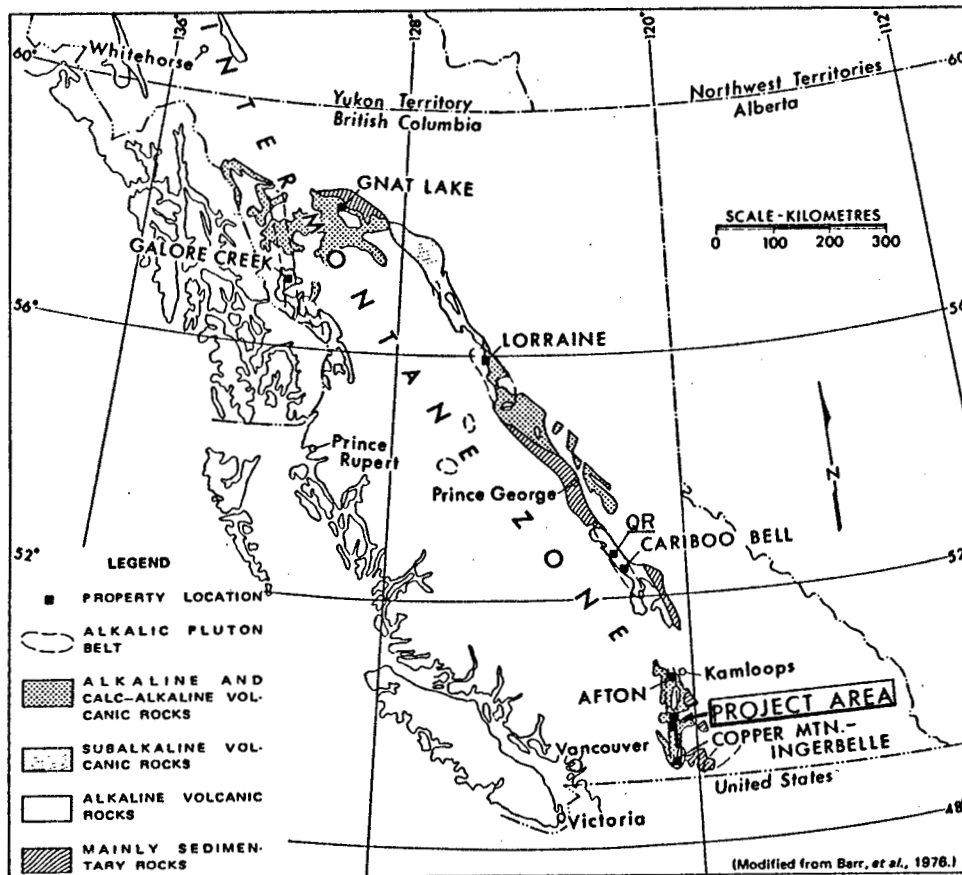
A preliminary diamond drilling programme was laid out to test the quartz-vein stockwork exposed by the trenching programme. Six vertical NQ diameter holes totalling 292 metres were drilled by Rainbow Diamond Drilling Ltd. of Merritt. The holes were evenly spaced along a 200-metre strike length; maximum depth attained was 94 metres in hole 87-3. The planned programme was curtailed by broken ground, and particularly by a wide easterly dipping shear zone. Circulation losses and high compressive forces prevented four of the holes penetrating the shear and only one of the other holes (87-3) reached target depth. All core was logged, split and sampled at one metre intervals. 270 samples were shipped to Acme Analytical Laboratories where they were analysed for gold, silver, lead and copper.

The split core is stored in covered racks at the Willow Heights Ranch, Aspen Grove.

GEOLOGY

Regional

The Upper Triassic Nicola Group rocks extend from the 49th parallel north to Kamloops Lake, and continue north beneath Tertiary cover to emerge in the Quesnel area as the Quesnel Belt (Preto, 1979).



Upper Triassic and Lower Jurassic volcanic rocks, significant copper deposits, and associated alkalic plutons in the Intermontane Zone.

LARAMIDE RESOURCES LTD.

SADIM PROPERTY

REGIONAL GEOLOGY

SCALE 1:13,500,000(approx) MARCH 1987

FIG. 3

I.M. Watson & Associates Ltd.

The volcanics of the Quesnel and Nicola Belts form a mixed alkaline and calc-alkaline sequence of basalts and derived breccias, tuffs, and minor sediments.

The volcanic rocks are intruded by comagmatic alkaline plutons, ranging in composition from syenogabbro to alkali syenite. The intrusions appear to be structure related and occur in belts along major lineaments and faults. They vary in size from plugs to small batholiths, and have been emplaced into the volcanic centres which produced the abundance of volcanic material (Barr et al, 1976).

In the Allison Lake-Missezula area, Preto has delineated three assemblages - a Western Belt of easterly dipping calc-alkaline flows, pyroclastics and sediments; a Central Belt of alkaline and calc-alkaline volcanics and intrusions, and minor sediments; and an Eastern Belt of westerly dipping volcanic sediments, tuffs and alkaline flows associated with small monzonite porphyry stocks. The belts are separated by major north-striking faults.

Preto believes that the Central Belt of dominantly volcanic rocks originates from eruptive centres along the major fault system, and points out the greater concentrations of mineral deposits along this belt.

The SADIM claims lie immediately west of the Summers Creek Fault, which marks the eastern boundary of Preto's Central Belt.

The property is underlain by northerly striking intermediate to basic flows, green monolithic and polyolithic volcanic breccias, tuffs, and less abundant argillites and limestones. These rocks have been intruded by irregular bodies of gabbroic to dioritic composition. Volcanics and sediments marginal to the intrusions have been variably propylitised (epidote-pyrite-chlorite-carbonate) and locally host erratically distributed copper-pyrite zones.

SADIM Gold Showing Area (Figures 5, 6a-c, 8, 9 and 10)

Lithology and Structure: The trenching and drilling programmes have provided additional detailed information regarding the lithology, structure and mineralisation of

the area hosting the SADIM gold zone. The geological setting is shown on the 1:2500 geological plan (Figure 5).

The general trend is slightly west of north; dips are steep to moderate easterly. 'Tops' have yet to be recognised.

A major easterly dipping shear zone was intersected in all the drill holes and projects to surface along the north trending swamp in the middle of the map area (Figures 5, 8, 9 and 10). The fault, probably a thrust, separates essentially andesitic flows (**1a**) and tuffs (**1e**) on the west from mixed tuffs (**1e**, **1e sil**, **1e cal**) on the east. The fault zone, which is about 15 metres thick, occurs along a dark grey carbonaceous limestone (See Figures 8 and 10), but also contains thick sections of quartz rich gouge.

The shear has caused intense and extensive fracturing and alteration (silification, pyritisation) in the adjacent rocks, particularly the tuffs above and to the east of the fault.

The silicified tuffs (**1e sil**) appear to be altered equivalents of the green and grey tuff (**1e**)-contacts in core and outcrop are irregular and transitional, and veins within the darker tuffs have alteration 'haloes' along their contacts. The silicified tuffs are pale grey, fine grained, and contain abundant small closely packed quartz eyes in a fine feldspathic groundmass. Pyrite occurs as fine disseminations, and is concentrated most heavily in zones of veining and fracturing. Weathering has oxidised the pyrite and the tuffs are strongly and pervasively hematitised to a depth of 10 metres.

An extensive fracture controlled quartz vein stockwork has developed in the altered tuffs, particularly in the silicified tuffs. Veins range from hair fractures to greater than one metre in thickness. There appear to be two dominant strike directions, roughly 30° north and south of east-west. Dips are southerly; the diagrammatic representation of veins in the 3+75S cross section (Figure 10) suggests that dips tend to flatten with depth as the veins close with the major shear zone.

Mineralisation: The quartz veins contain erratically disseminated sulphides, mainly pyrite, as well as chalcopyrite, and less commonly galena. In many cases, sulphides are concentrated along the vuggy margins or centres of a vein. Galena is usually

present as very fine crystal clusters or linings along hair fractures in the quartz. Sulphide concentration is related to vein size and to density of fracturing of the host tuff. Trench and drill hole sampling results show a close relationship between precious metal content, quartz veining (and fracturing), and sulphide concentration. The presence of galena is a good indication of elevated gold and silver content.

SUMMARY OF RESULTS

1. Gold mineralisation on the SADIM property occurs in a quartz-vein stock work within silicified and calcareous tuffs (**le sil** and **le cal**) above a major northerly striking, easterly dipping shear zone.
2. Significant precious metal contents have been obtained from trench sampling over an area approximately 200 metres by 60 metres. The gold content of chip samples ranges from 50 to 4,350 ppb. A 1.1 metre vein in Trench #2 assayed 6,390 ppb Au (0.19 opt.).
3. The six hole, 292 metres preliminary drilling programme was curtailed by bad ground within and adjacent to the shear zone, but tested the stockwork to a depth of 50 metres (Hole 87-1). Drill hole and trench sample data suggest that precious metal content increases from south to north (Figure 8a). The most northerly hole, 87-6, cut a 9.0 metre section from 22 metres to the bottom of the hole which averages 3,090 ppb Au (0.09 opt.) and 25.4 ppm Ag (0.74 opt.) and includes a one metre section assaying 19,800 ppb Au (0.58 opt.) and 159.1 ppm Ag (4.6 opt.).
4. The stockwork is open to the east (down-dip) and to the north. Although gold and silver content appears to diminish to the south, further, deeper drilling is needed to establish the southern limits of the mineralised veins.
5. Comparison of the geological and assay plans and sections shows a close correlation of gold/silver content with quartz veining and sulphide concentration (pyrite, chalcopyrite and galena in order of abundance). The presence of galena is a good indication of elevated gold and silver content. The Au:Ag ratio is consistently 1:8.

6. Further drilling and trenching is required to establish the full extent and tenor of the SADIM stockwork. The northerly strike and easterly down-dip potential of the zone are obvious priority targets.

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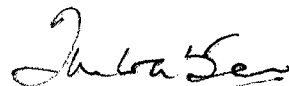
I.M. Watson, P.Eng.

CERTIFICATE OF QUALIFICATIONS

I, Ivor Moir Watson, of 584 East Braemar Road, North Vancouver, British Columbia, hereby certify that:

1. I am a consulting geologist with offices at 816 - 675 West Hastings Street, Vancouver, B.C.
2. I am a graduate of the University of St. Andrews, Scotland (B.Sc. Geology 1955).
3. I have practised my profession continuously since graduation.
4. I am a member in good standing of the Association of Professional Engineers of B.C., and a Fellow of the Geological Association of Canada.
5. Work on the SADIM Property was carried out during the periods September 23rd to October 2nd, 1986; December 10th to 17th, 1986; and January 24th to February 11th, 1987 by the following personnel:
 - I. M. Watson - Geologist/Supervisor
 - J. H. Randa - Prospector/Sampler - Sept. 23 to Oct. 2,
 - D. England - Sampler - Sept. 29 to Oct. 2, 1986
 - D. Whalen - Prospector/Sampler - December 10 to 17, 1986
 - S. Angus - Prospector/Sampler - December 10 to 17, 1986
 - R. Gibbs - Core Splitter - January 24 to February 11, 1987

March 26, 1987
Vancouver, B.C.



I.M. Watson, B.Sc., P.Eng.

REFERENCES

Barr, D.A., Fox, P.E., Northcote, K.E., and Preto, V.A., 1976. The Alkaline Porphyry Deposits - A Summary; in CIM Special Vol. No. 15.

Preto, V.A., 1975. Notes to Accompany Preliminary Map No. 17. Geology of the Allison Lake - Missezula Lake Area. B.C. MEMPR.

1979. Geology of the Nicola Group between Merritt and Princeton, Bull. 69, B.C. MEMPR.

Watson, I.M., 1985. Reconnaissance Geological and Geochemical Surveys of the SADIM Group, for Laramide Resources Ltd.

Assessment Reports

#517 - 1963 Report on the K.R. Group of Plateau Metals Ltd. by Asarco Smelting & Refining Co. (Geology, magnetometer survey.)

#985 - 1967 Geochemical report on the K.R. Group by C. Lammle for Adera Mining Ltd.

#3363 - 1971 Geological, Geochemical and Geophysical Report on the Ketchan Creek property by J. Christofferson, G. DePaoli, and C. Hodgson for Amax Exploration Inc.

#5044 - 1973 Geological and Prospecting Reports on the Cindy Group by D.C. Malcolm and E. Sleeman.

#6036 - 1976 Geochemical Report on Rum Claim Group by D.G. Mark for Ruskin Developments Ltd.

#8352 - 1980 Ground Magnetic and Soil Geochemical Survey over part of the Rum Property, by D.T. Mehner for Cominco Ltd.

#9407 - 1981 Soil Geochemical Survey over part of the Rum Property, by D.T. Mehner for Cominco Ltd.

Statement of Costs - SADIM Claims

Phase I Trenching, Sampling & Mapping Programme -September
23 to October 2, 1986
SADIM I - 5 Claims

Phase II Trenching, Sampling & Mapping Programme - December
10 - 17, 1986
SADIM I-4 Claims

Diamond Drilling Programme January 24 to February 11, 1987
SADIM I - 6 Claims

STATEMENT OF COSTS - SADIM CLAIMS

Phase I

a) Trenching - 23 - 26 September, 1986

Salaries

I.M. Watson (Consulting Geologist/Supervisor)
4 days @ \$400/day \$ 1,600.00

J. Randa (Prospector)
3 days @ \$185/day 555.00 \$ 2,155.00

Accommodation/Board 245.00

Vehicle Rental - (4 x 4)
4 days @ \$30/day 120.00

Fuel 51.50

Excavator Rental
JSWBH 90 Excavator
Operator
Mob & Demob

2,300.89 \$ 4,872.37

b) Geological Mapping/Sampling - 27 September to 2 October, 1986

Salaries

a) Field Work

I.M. Watson (Consulting Geologist/Supervisor)
6.5 days @ \$ \$400/day \$ 2,600.00

J. Randa (Prospector)
6.5 days @ \$85/day 1,202.50

D. England (Labourer/Assistant)
3 days @ \$116/day 348.00 5,150.50

b) Report Preparation

I.M. Watson
3 days @ \$400/day 1,200.00

Accommodation/Board 660.31

Telephone, Freight 49.96

Vehicle Rental
6 days @ \$30/day 180.00

Fuel 50.75

Supplies 80.67

Geochemical Analyses - Acme Laboratories
186 samples @ \$11.50/ea (Au,Ag,Pb,Cu) 2,147.07

Drafting - D. Phillips
12 hrs. @ \$20.00/hr. 240.00 8,789.26

Total Phase I \$13,631.63

Phase II

a) Trenching Dec. 10-15, 1986

Salaries			
I.M. Watson (Consulting Geologist/Supervisor)			
3 days @ \$400.00/day		\$ 1,200.00	
Accommodation/Board		105.00	
Vehicle Rental (4 x 4)			
3 days @ \$30.00/day		90.00	
Fuel		22.98	
Excavator Rental			
JSWB490 - 41.5 hrs. @ \$95.00	\$ 3,942.50		
Operator Costs			
16.5 hrs. @ \$20.00/hr.	330.00		
Mob/Demob - 7 hrs. @ \$66.75	467.25	4,739.75	6,157.73

b) Geological Mapping/Sampling Dec. 11-17, 1986

Salaries			
I.M. Watson (Consulting Geologist/Supervisor)			
5 days @ \$400.00/day	2,000.00		
S. Angus (Prospector/Sampler)			
8 days @ \$140.00/day	1,120.00		
D. Whalen (Prospector/Sampler)			
8 days @ \$180.00/day	1,440.00	4,560.00	
Accommodation/Board		840.09	
Telephone, Freight		34.94	
Vehicle Rental (4X4)			
5 days @ \$30.00/day		150.00	
Fuel		45.97	
Supplies		199.84	
Geochem Analyses (Acme Analytical Labs)			
Au, Ag, Pb, Cu (175 samples + freight)		2,138.14	
Drafting - D.L. Phillips Drafting Services			
22 hrs. @ \$20.00/hr.		440.00	8,408.98
			<u>8,408.98</u>
	Total Phase II		<u>\$14,566.71</u>

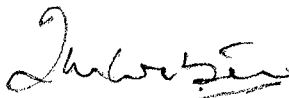
Diamond Drilling Programme - 24 January - 11 February, 1987

Salaries		
I.M. Watson (Consulting Geologist/Supervisor)		
19 days @ \$400.00/day	7,600.00	
R. Gibbs (Core Splitter)		
19 days @ \$115.00/day	<u>2,185.00</u>	9,785.00
Accommodation/Board		1,802.31
Telephone/Freight		126.82
Vehicle Rental (4 x 4)		
19 days @ \$35.00/day		665.00
Fuel		269.46
Supplies/Equipment		686.86
Diamond Drilling - Rainbow Diamond Drilling		
292 m NQ @ \$59.05/m (1,800 ft.)	17,244.00	
Filed costs, consumables	<u>5,661.00</u>	22,905.00
Assaying - Acme Analytical Laboratories		
270 core samples @ \$10.50/ea	2,835.00	
Shipping	<u>96.98</u>	2,931.98
Map Printing		163.14
Drafting - D.L. Phillips Drafting		
41.5 hrs. @ \$20.00/hr.	<u>830.00</u>	<u>40,105.57</u>
	Total	<u>\$40,105.57</u>

SUMMARY

Phase I Trenching, mapping, sampling	\$ 8,759.26
Phase II Trenching, mapping, sampling	14,566.71
Diamond Drilling	<u>40,105.57</u>
	<u>\$63,431.54</u>

I.M. WATSON & ASSOCIATES LTD.


I.M. Watson, P.Eng.

APPENDIX A
DIAMOND DRILL LOGS

DIAMOND DRILL RECORD

PROPERTY Sagin

HOLE NO. 87-1 P3

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS			
						Gr ppb	Ag ppm	Pb ppm	Cu ppm
(7.00-28.8)	Vn. CONTAINS TWO GENERATIONS Qtz - Don. CRACKS, CUT?								
Cont'd.	By MILKY, ORANGE. Pyrite IN Vn (Wk.) & WALL ROCK (MEAN)								
	9.60-9.90: Vnlts. ~ 75%CA + Mod. Dissem. Py								
	10.5-11.8: 7%CA. MINOR Py IN Vn., ONLY DISSEM. IN WALL ROCK.	88708	10.0	11.0	1.0	31	0.1	4	131
	12.0-12.5: Vnlts. ASSOCI'D WITH CHL SLIPS 15%CA	88709	11.0	12.0	1.0	355	2.2	6	57
	13.0-13.3: " " " " 30%CA								
	13.6-14.2: ABUND. F. Vnlts IN SHATTERED (BUT TIGHT) TUFF ASSOCI'D. WITH SPARSELY DISSEM. Py. Vn. @ 14.2-20%CA.	88710	12.0	13.0	1.0	10	0.1	3	18
	CONTAINS Py, CCP (MILKY MARGINS) 16.7: Vn @ 60%CA (BROKEN)	88711	13.0	14.0	1.0	265	1.5	6	58
	17.0-17.6: NUMEROUS RANDOMLY ORIENTED VnS/Vnlts (BROKEN CORE) ASSOCI'D WITH CHL LINED FRACTS @ 30%CA.	88712	14.0	15.0	1.0	995	6.2	5	51
	17.7: 55%CA (~ 0.5cm), SL. VUGGY.								
	18.3: ROMICRYNE, ENCL. SMALL EX. FRAGS.	88713	15.0	16.0	1.0	52	0.3	2	45
	18.7: 40%CA (~ 3cm) Py ALONG CONTACTS								
	19.0 & 19.1: 55%CA (1cm) - ASSOCI'D WITH CHL SLIPS (19.1-20.2)	88714	16.0	17.0	1.0	33	0.1	4	4
	20.9-20.3: Qtz Vn + Py 'ZONE' - Vn @ 20.9; 21.4-21.5;	88715	17.0	18.0	1.0	12	0.1	6	51
	21.6-22.0 25%CA INCL. Py, CCP, v.f. PDS								
	22.1-22.2: Vnlts @ 50%CA.	88716	18.0	19.0	1.0	185	0.9	7	83

DIAMOND DRILL RECORD

 PROPERTY SADIM

 HOLE NO. 87-1 87

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS			
						Pb ppm	Ag ppm	Pb ppm	Cu ppm
39.2-39.5 (0.3m)	ALTERATION ZONE (INTERVAL UNCERTAIN - GROUND CORE) CASBY XTALINE QZTQ - FELSPATHIC, FINELY FRACT. CML / SEC. DEVELOPED ALONG FRACT PLANS. CONTAINS FINELY & WILLY DISSESS. Py. (CORE RECOVERY ~ 50%?)	88737	39.0	40.0	1.0	109	0.8	4	60
39.5-39.7 (0.2m)	TUFF SILICIFIED - AS ABOVE - BROKEN FINELY & WILLY FRACTURED								
39.7-41.0 (1.3m)	ALTERATION ZONE (AS SEEN 39.2-39.5 ABOVE) PALE GREEN - WHITE, MINOR CML ALONG FRACT. PLANES - CONTAINS FRAGS. OF TUFF SILICIFIED TUFF. MINOR FINELY DISSESS. Py. (CORE RECOVERY ~ 65%)	88738	40.0	41.0	1.0	151	1.1	2	44
41.0-41.5 (0.5m)	FAULT - 10cm. GYROGNE ZONE, QZTQSE.	88739	41.0	42.0	1.0	285	1.9	7	33
41.5-42.2 (0.7m)	TUFF SILICIFIED? - MAROON - RED BROWN QZTQSE (TIGHTLY BLENDED) SMALL QZTQSE, FINELY HEAVILY & TIGHTLY FRACT., FRAGS. HEAVILY BY QZTQ VENTS	88740	42.0	43.0	1.0	6	0.1	7	29

DIAMOND DRILL RECORD

 PROPERTY SADIM

 HOLE NO. 87-1 P8

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS			
						As pct	Ag ppm	Pb ppm	Cu ppm
42.2-42.7 (0.5m)	TUFF 'SILICIFIED', AS ABOVE, FINELY FRACT., SHEARED - NUMEROUS V. FINE CONTACTED QTZ VENTS.								
42.7-43.3 (0.6m)	TUFF: MARLON / BUFF. INTENSELY SHEARED & FRACT., FINELY Banded appearance								
43.3-51.51 (8.2m)	FAULT ZONE: Major Shear / Fault (Bredden) DOMINANTLY QTZ GREY / WHITE. UPPER 10cms. Gouge. SECTION INTENSELY SHEARED, INTENSITY INCR. DOWNHOLE UNTIL ROCK BECOMES QTZ GOUGE BY FINELY DISSEMIN. THROUGHOUT, STRONGEST IN UPPER PORTION TO 46.3m - QZ LARGES THAN FURTHER ALONG QTZ FRAGS - ELONGATIONS OF FRAGS. IN DIRECTION OF SHEARING 15-30°/CA VAR. 44.0 : SEM. GOUGE ZONE 46°/CA 44.8 : SHEARS @ 10°/CA	88741	43.0	44.0	1.0	275	2.8	11	63
		88742	44.0	45.0	1.0	355	4.8	77	20
		88743	45.0	46.0	1.0	91	5.9	28	223
		88744	46.0	47.0	1.0	845	9.5	4.9	106
		88745	47.0	48.0	1.0	615	5.6	67	80
	45.4-46.3 - MILKY & GREEN GREY QTZ., F. WR. PY. 46.33-47.85 - GOUGE. QTZ., FINELY HYDRATISED 47.85-49.1 - TUFF, PILE GREY. BUFF. HIGHLY ALT. SMO & FRACT.	88746	48.0	49.0	1.0	6	0.1	29	29

CLAIM NO. _____

DIAMOND DRILL RECORDPROPERTY SADOMHOLE NO. 87-3 D₁LATITUDE 4 100 S

ELEVATION _____

BEARING -DEPTH 93.6 mSTARTED 2.2.87COMPLETED 5.2.87

CORE: NO

DEPARTURE 7+78.6 W

SECTION _____

DIP -90°DRILLED BY RANSOM DIAMOND DRILLINGLOGGED BY INW

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS			
						Au ppm	Ag ppm	Pb ppm	Cu ppm
0 - 3.05	CASING								
3.05-12.4 (9.35m)	TUFF: SILICIFIED - DOM. POLE GREY TUFF. QTZ EYE TUFF, HEAVILY KENAFITE STAINED, & HIGHLY BUT FINELY FRACTURED - DOM. FRACT. DIR. SO-SS/CA OCASIONAL ZONES OF 'UNALTERED' FINE GRAY TUFF ex. 3.73-4.0; 5.2-5.4; & 5.7-5.8. RUSTY IRON. DISPERSED THROUGHOUT, FINE. STRONGEST FRAGMENT QTZ VNS/VNDS. QUARTZ VNS:- 7.2 : (0.2 cm) 7.7-7.9 : ~ 5°/CA 8.4-8.7 : 40°/CA Py IN IRON STRONGEST // VN. 9.1-9.2: 10.0 : (0.5 cm) 10.2-10.3: Bx'D 40°/CA 11.5-11.6: Bx ZONE CONTAINING F. DR. METALLIC? STRUCTURE - RUSTY NARROW GORGE ZONES @ 6.9 (60°/CA)	88787	3.05	4.0	0.95	1	0.1	7	8
		88788	4.0	5.0	1.0	1	0.1	2	74
		88789	5.0	6.0	1.0	1	0.1	2	47
		88790	6.0	7.0	1.0	1	0.1	4	46
		88791	7.0	8.0	1.0	82	0.6	9	105
		88792	8.0	9.0	1.0	97	0.3	5	83
		88793	9.0	10.0	1.0	99	17.3	4	86
		88794	10.0	11.0	1.0	1	0.1	4	56

DIAMOND DRILL RECORD

 PROPERTY SADIN

 HOLE NO. 87-3 13

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS			Gt/ton						
						An. pct	Ag. pct	Pb. pct							
(14.6-40.1)	ARE IN & ADJACENT TO QTZ. VNS / FRACCS	88799	15.0	16.0	1.0	3620	37.1	1526	150						
Cont'd	QUARTZ VEINS / SULPHIDES -	88800	16.0	17.0	1.0	40	0.10	2	59						
	15.5-15.85 : QV HEAVY DISSEM Pbs, QV. CEP	88951	17.0	18.0	1.0	1	0.3	8	73						
	20.25-20.37 : QV MINOR Pbs, HEAVY DISSEM P ₄ @ F.W. CONTACT	88952	18.0	19.0	1.0	1	0.2	6	20						
	(30°/CA)	88953	19.0	20.0	1.0	1	0.3	2	65						
	20.5 : QV : (0.5cms) 25°/CA	88954	20.0	21.0	1.0	87	1.1	8	98						
	27.5-27.6 : P ₄ , MINOR FINELY DISSEM.	88955	21.0	22.0	1.0	1	1.3	4	48						
	28.5-28.6 : QV 70°/CA, TR. V.F. SULPHIDES	88956	22.0	23.0	1.0	1	0.2	10	30						
	30.0 - DOWNHOLE INCREASING PYRITE, V.F. MOD.	88957	23.0	24.0	1.0	1	0.2	2	78						
	- HEAVILY DISSEM IN TIGHT ZONES ALONG QTZ VNS/VEINS	88958	24.0	25.0	1.0	65	0.4	6	52						
	31.0 : QV (3cms) 65°/CA	88959	25.0	26.0	1.0	1	0.1	5	19						
	31.5 : QV (2cms) 50°/CA - P ₄ MOD. DISSEM	88960	26.0	27.0	1.0	155	1.3	8	445						
	30.8-31.4	88961	27.0	28.0	1.0	106	1.0	4	231						
	31.9-32.4 : P ₄ , MOD. DISSEM	88962	28.0	29.0	1.0	92	0.4	6	74						
	32.2 : QV (1cm) 40°/CA	88963	29.0	30.0	1.0	19	0.1	2	76						
	32.6-32.9 : QTZ VNS 55°/CA, P ₄	88964	30.0	31.0	1.0	1	0.1	4	102						
	33.9 : QV 70°/CA P ₄	88965	31.0	32.0	1.0	350	2.6	10	77						
	34.4 : QV 35°/CA P ₄	88966	32.0	33.0	1.0	155	1.3	5	120						
	37.9-38.4 ~ 5-10% P ₄ .	88967	33.0	34.0	1.0	17	0.1	6	54						
	38.8-40.1 ~ HEAVILY DISSEM ALONG FRACCS (~30°/CA)	88968	34.0	35.0	1.0	195	1.4	8	126						

DIAMOND DRILL RECORD

 PROPERTY SADIM

 HOLE NO. 87-3 P6

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS				
						Ag ppb	Ag ppm	Pb ppm	Cu ppm	Zn ppm
48.7-49.1 (0.4m)	LIMESTONE - DK. GREY - BLACK, V.F. CR - IN PART CLOUDY BANDED @ 50°/CA. SPARSELY DISSEM. BY 1/2 MINUTE CUBES LOWER CONTACT @ STRONG SLIP (1cm. GORGE) @ 70°/CA CORE BROKEN - CORE RECOVERY 50%?	88979	48.0	49.0	1.0	5	0.1	29	16	
49.1-49.95 (0.85)	CONCRETEOUS GRTS - GREY - BUFF. ELONGATED SMALL LST? CLOTS. FINE COARSE FOLIN @ 50°/CA MINOR SMALL QZ/CARB? YLTS., NO OBL. PREFERRED ORIENTATION FRAGS. DARK LST @ LR. CONTACT (67°/CA FOLIN) RECOVERY 90%	88980	49.0	50.0	1.0	2	0.3	13	25	49
		88981	50.0	51.0	1.0	1	0.3	7	46	57
49.98-51.55 (4.57m)	CONCRETEOUS APRILL / LIMESTONE: PALE GREY, V.F. CR. UNIFORM, NEAR MASSIVE IN PART, BUT GENERALLY WEAKLY FOLD'D., FINELY FRACTURED. MOD-WKLY CARC. PYRITE: SPARSE, SMALL XTALS: SCATTERED CARB. YLTS (FRACTURE HEALING) 50.9: GORGE (3cm) @ 60°/CA 52.4 & 53.2 FOLIN @ 65°/CA CORE RECOVERY 75%	88982	51.0	52.0	1.0	13	0.6	7	51	57
		88983	52.0	53.0	1.0	11	0.4	9	60	58
		88984	53.0	54.0	1.0	9	0.1	7	45	46

DIAMOND DRILL RECORD

 PROPERTY SADIM

 HOLE NO. 87.3 Pg

DEPTH	FORMATION	SAMPLE NO.	FROM	TO	WIDTH	ASSAYS				
						Au ppm	Ag ppm	Pb ppm	Cu ppm	Zn ppm
66.4-68.75 (2.35m)	TUFF SILICIFIED (POSS INTERBED AND. TUFF) DOM. BUFF V.F.GR., LOCALLY HOSTED PURPLE-GREEN INCL. NARROW ZONES OF DK. GREEN AND TUFF	88997	66.0	67.0	1.0	1	0.1	6	76	74
66.9-67.3	QTZ HEALED FRACTS @ 40-60°/CA NARROW (1cm), WIDELY FURTHERED	88998	67.0	68.0	1.0	3	0.1	6	74	71
67.6	5cm. Gouge Zone 60°/CA RECOVERY 95%+	88999	68.0	69.0	1.0	33	0.1	7	87	67
68.75-70.15 (1.40m)	FAULT GouGE. GRAY-GREEN, INCL. FRACTS. BUT ALT. SEC. AND TUFF? UPPER CONTACT SHARD @ 55°/CA MINOR QTZ. AS FRACTS IN LOWER 0.2m OF GouGE. SCATTERED V.F. PY XTALS IN GouGE RECOVERY 85%+	89000	69.0	70.0	1.0	4	0.8	25	76	285
70.15-76.0 (5.85m)	ANDESITE BRECCIA. DOM. DK. GREEN CONT. GREEN & PURPLE FRACTS UP TO 1cm IN TUFFACEOUS CLAYMASS. ABUND. FRACTS. QTZ (CARB?) HEALED, - BLEACHED ZONES FRINGING QTZ VNS/FRACTS.	88851	70.0	71.0	1.0	2	0.1	6	98	74
		88852	71.0	72.0	1.0	3	0.1	6	75	67

APPENDIX B
GEOCHEMICAL ANALYTICAL REPORTS

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: SEPT 29 1986

DATE REPORT MAILED: *Oct 8/86*

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: ROCK CHIPS AU* ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Toy*...DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON & ASSOCIATES

FILE # 86-2930

PAGE 1

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2129	84	11.4	1290
2130	66	7.7	850
2131	29	6.8	820
2132	7	1.6	225
2133	6	1.3	185
2134	8	.4	52
2135	1105	26.6	2940
2136	16	5.7	810
2137	27	8.6	1310
2138	21	1.0	165
2147	132	13.2	2350
2148	29	2.7	310
2149	32	8.7	1090
2150	31	7.6	930
016101	2191	16.0	1790
STD C/AU-R	41	7.1	510

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: OCT 4 1986

DATE REPORT MAILED: *Oct 10/86*

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.NG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: ROCK CHIPS AU ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Toy* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I. M. WATSON

PROJECT-SADIM FILE# 86-3055

PAGE 1

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2107	41	6.0	680
2108	13	12.5	1250
2109	82	2.5	220
2110	123	8.7	875
2111	15	.4	24
2112	24	1.5	127
2113	10	.4	24
2114	13	1.7	210
2115	27	28.1	3450
2116	7	2.1	305
2117	8	.3	38
2118	6	.3	31
2119	19	14.0	1510
2120	8	1.2	85
2121	8	4.9	730
2122	5	20.0	39
2123	3	.9	93
2124	7	4.7	665
2125	4	.1	10
2126	4	.3	13
2127	7	.3	2
2128	8	.6	6
2139	13	.2	12
2140	17	.1	19
2141	11	.3	36
3601	11	.8	110
3602	37	1.4	195
3603	39	22.5	2700
3604	9	.6	61
3605	17	13.3	1810
3606	109	15.6	1860
3607	7	.3	22
3608	22	7.8	1030
3609	1990	56.2	6470
3610	1269	11.4	1130
3611	19	2.7	360
STD C/AU-R	38	6.7	510

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
3612	83	3.8	450
3613	51	15.9	2080
3614	6	6.9	1090
3615	18	6.6	1060
3616	2927	56.4	6390
3617	21	14.6	2190
3618	65	18.7	2180
3619	29	3.8	550
3620	1566	31.4	3450
3621	175	14.3	2060
3622	1468	41.2	4590
3623	40	5.4	810
3624	647	25.0	3110
3625	4	4.4	1060
3626	8	1.3	240
3627	15	2.3	460
3628	121	6.9	950
3629	344	18.2	2100
3630	141	8.1	1090
3631	40	10.8	2100
3632	9	6.5	920
3693	7	13.2	1850
3694	55	14.6	1600
3695	6	10.3	1620
3696	57	23.4	2960
3697	16	6.5	850
3698	5	1.3	190
3699	3	13.8	1610
3700	82	25.0	3200
3751	6	3.2	390
3752	8	5.5	930
3753	22	9.7	1380
3754	3	6.8	930
3755	2	2.9	420
3756	7	.9	93
3757	8	9.8	1490
STD C/AU-R	37	6.8	505

I.M. WATSON

PROJECT-SADIM

FILE# 86-3055

PAGE 3

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
3758	2	2.2	215
3759	6	2.0	270
3760	26	4.7	490

ACME ANALYTICAL LABORATORIES LTD.
352 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: OCT 4 1986

DATE REPORT MAILED: *Oct 16/86*

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: ROCK CHIPS AU ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Toye* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON

PROJECT-SADIM FILE# 86-3057

PAGE 1

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2142	8	2.2	290
2143	10	.2	14
2144	10	.5	36
2145	51	7.0	720
2146	120	2.7	260
3651	12	.5	35
3652	4	.2	1
3653	7	.2	9
3654	7	.2	5
3655	7	.2	9
3656	10	9.6	1310
3657	5	.8	160
3658	2	.2	16
3659	29	8.9	1160
3660	7	.5	15
3661	9	2.0	290
3662	9	.3	14
3663	11	.2	4
3664	11	.6	25
3665	8	.2	12
3666	11	.3	8
3667	8	.5	44
3668	13	2.9	380
3669	1001	19.3	2350
3670	414	14.9	1550
3671	168	9.3	1080
3672	17	1.0	123
3673	13	.5	45
3674	11	.3	14
3675	19	1.0	95
3676	21	1.9	240
3677	15	.2	12
STD C/AU-R	39	7.2	490

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
3678	15	.4	12
3679	20	.7	10
3680	25	1.2	93
3681	4	.1	2
3682	3	.1	1
3683	3	1.0	13
3684	3	.7	2
3685	2	.2	1
3686	5	.3	3
3687	4	.5	4
3688	11	.5	7
3689	9	.2	6
3690	10	.1	8
3691	9	.2	12
3692	10	.3	17
16102	18	9.5	1150
16103	30	2.7	290
16104	49	8.1	1150
16105	114	16.9	2250
16106	1472	34.3	4350
16107	12	1.2	106
16108	20	1.4	67
16109	11	3.2	950
16110	4	.3	21
16111	7	.3	5
16112	7	1.7	215
16113	9	.2	4
16114	8	.6	31
16115	10	.3	8
16116	8	1.3	112
16117	9	.2	9
16118	18	1.4	86
16119	8	.2	16
16120	5	.3	6
16121	7	.2	4
16122	6	.5	42
STD C/AU-R	38	7.0	520

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
16123	335	10.1	1180
16124	271	9.2	1240
16125	10	.9	106
16126	10	.5	57
16127	224	10.1	1450
16128	450	17.0	2080
16129	7	.5	118
16130	44	2.7	290
16131	7	5.7	840
16132	14	3.6	480
16133	7	.2	32
16134	9	.5	26
16135	6	1.5	210
16136	4	.5	56
16137	12	2.7	340
16138	14	1.5	159
16139	36	8.1	1150
16140	9	4.5	680
16141	9	4.4	630
16142	11	8.3	1300
16143	15	3.8	405
16144	11	1.8	160
16145	10	1.9	350
16146	6	.4	50
16147	10	3.3	540
16148	12	4.9	860
16149	7	6.8	1250
16150	5	2.1	149
STD C/AU-R	40	6.9	480

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: DEC 3 1986

DATE REPORT MAILED: Dec 8/86....

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: SOILS -80MESH AU* ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. J. J.* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON & ASSOCIATES FILE # 86-3894 PAGE 1

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
TR#2 12W 0.5-1.5	22	.7	30
TR#2 12W 1.5-2.5	18	.9	84
TR#2 12W 2.5-3.0	11	1.3	290
TR#2 12W 3.0-3.25	24	5.3	445
TR#2 12E 0-0.5	35	1.4	149
TR#2 11W 0-1	15	1.9	106
TR#2 11W 1-1.5	26	2.6	119
TR#2 11W 1.5-1.75	6	.3	99

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: DEC 23 1986

DATE REPORT MAILED: *Jan. 13/87...*

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: ROCK CHIPS AU ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Toye* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON PROJECT-SADIM FILE# 86-4071 PAGE 1

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2421	8	.1	22
2422	15	.3	25
2423	47	3.0	380
2424	145	4.7	845
2425	23	3.3	520
2426	22	5.0	680
2427	25	2.7	410
2428	9	2.9	495
2429	177	7.0	960
2430	12	2.3	350
2431	7	.6	49
2432	9	.3	18
2433	7	.7	66
2434	3	.2	42
2435	8	.2	29
2436	9	.1	9
2437	8	.1	6
2438	12	.1	10
2439	5	.1	14
2440	6	.1	7
2441	6	.1	11
2442	6	.1	8
2443	4	.5	48
2444	10	.3	16
2445	9	.3	34
2446	8	.1	12
2447	8	.3	19
2448	11	.2	9
2449	9	.4	25
2450	12	.2	10
2507	17	1.2	22
2508	41	7.5	435
2509	9	2.2	320
2510	12	1.5	180
2511	7	.1	27
2512	7	2.3	510
STD C/AU-R	39	6.9	530

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2513	8	2.8	320
2514	6	.2	7
2515	7	.2	6
2516	5	.2	3
2517	6	.1	4
2518	5	.2	5
2519	9	.4	75
2520	11	.1	10
2521	9	8.0	1430
2522	11	25.9	144
2523	2377	76.7	9140
2524	18	.2	17
2525	44	.6	36
2526	39	.4	15
2527	21	.3	3
2528	10	.1	56
2529	10	.1	2
2530	14	.2	4
2531	12	.2	18
2532	10	.1	1
2533	9	.1	1
2534	6	1.2	140
2535	59	16.8	1820
2536	5	.3	29
2537	5	.6	46
2538	5	.1	9
2539	5	.1	15
2540	7	.1	1
2541	5	.2	2
2542	8	.1	1
2543	7	.1	1
2544	7	.1	1
2545	4	.4	41
2546	8	.5	12
2547	3	.2	1
2548	127	40.4	4310
STD C/AU-R	38	7.0	480

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
2549	128	61.7	7110
3633	9	1.7	195
3634	12	3.8	565
3635	8	2.3	270
3636	4	1.2	190
3637	5	.3	24
3638	9	2.2	245
3639	6	.9	142
3640	6	1.2	133
3641	7	1.7	335
3642	3	1.3	119
3643	7	2.0	195
3644	4	.7	81
3645	8	.9	62
3646	3	.8	165
3647	7	1.9	255
3648	6	1.5	235
3649	5	.9	152
3761	4	.1	4
3762	7	.1	6
3763	6	.1	10
3764	8	.1	2
3765	8	.1	1
3766	5	.1	1
3767	6	.1	2
3768	9	.2	2
3769	44	.2	5
3770	43	.3	3
3771	7	.2	2
3772	7	.4	7
3773	9	.2	43
3774	6	.1	6
3775	4	.1	1
3776	6	.1	2
3777	6	.1	1
3778	10	.1	10
STD C/AU-R	38	6.9	505

SAMPLE#	Pb PFM	Ag PFM	Au* PPB
3779	4	.2	3
3780	8	.1	2
3781	4	.3	18
3782	4	.1	1
3783	2	.1	1
3784	4	.1	14
3785	2	.1	7
3786	5	.1	13
3787	9	.3	12
3788	104	18.2	2410
3789	4	2.0	390
3790	8	1.1	155
3791	5	.1	4
3792	4	.1	1
3793	6	.6	92
3794	567	50.8	5560
3795	21	5.1	780
3796	7	2.4	480
3797	931	62.9	7830
3798	12	17.2	3080
3799	8	5.2	1100
4005	9	.6	72
4006	2	2.0	330
4007	8	4.3	610
4008	2	1.3	280
4009	7	.9	65
4010	4	.1	10
4011	4	.1	1
4012	2	.1	3
4013	9	.1	4
4014	6	.1	6
4015	28	6.6	1160
4016	8	2.9	390
4017	10	1.3	152
4018	6	.3	23
4019	5	.2	8
STD C/AU-R	37	6.8	530

SAMPLE#	Pb PPM	Ag PPM	Au* PPB
4020	2	.4	8
4021	2	.2	6
4022	4	.1	16
4023	5	.2	2
4024	4	.3	3
4025	2	.3	14
4026	4	2.9	400
4027	9	1.9	240
4028	8	.5	21
4029	5	.5	13
4030	6	.4	19
4031	5	.5	8
4032	3	.4	60
4033	5	.5	71
4034	3	.1	2
4035	2	.1	28
4036	3	.2	1
4037	2	.2	3
4038	2	.1	1
4039	4	.1	12
4040	2	.4	1
4041	5	.2	8
4042	3	.1	1
4043	3	.1	3
4044	3	.3	2
4045	5	.1	7
4046	2	.2	1
4047	5	.2	6
4048	4	.1	27
4049	4	.1	1
4050	5	.1	1
STD C/AU-R	37	6.9	510

____ Assay required for correct result for Ag > 34 PPM

ACME ANALYTICAL LABORATORIES LTD.
 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
 PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: FEB 3 1987

DATE REPORT MAILED: Feb 6/87

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: CORE AU ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Deys* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON ASSOCIATES PROJECT-SADIM FILE # 87-0230 PAGE 1

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
088701	76	12	.8	105
088702	75	5	.1	11
088703	88	13	1.6	165
088704	205	350	7.9	890
088705	76	5	1.7	295
088706	108	7	3.1	445
088707	76	26	23.6	3260
088708	131	4	.1	31
088709	57	6	2.2	355
088710	18	3	.1	10
088711	58	6	1.5	265
088712	51	5	6.2	995
088713	45	2	.3	52
088714	41	4	.1	33
088715	51	6	.1	12
088716	83	7	.9	185
088717	67	2	1.4	205
088718	124	10	3.9	495
088719	106	173	18.1	2510
088720	98	6	2.6	475
088721	62	5	.9	195
088722	103	3	1.3	250
088723	328	11	10.4	1330
088724	174	4	5.2	740
088725	115	5	1.8	305
088726	167	6	3.0	435
088727	35	2	.1	19
088728	63	3	.3	43
088729	119	4	1.5	275
088730	83	12	3.2	505
088731	75	3	2.0	350
088732	114	2	2.3	335
088733	160	2	3.3	440
088734	46	2	.4	62
088735	100	2	1.0	215
088736	104	2	2.0	375
STD C/AU-R	61	41	6.9	490

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
088737	60	4	.8	109
088738	44	2	1.1	151
088739	33	7	1.9	285
088740	29	7	.1	6
088741	63	11	2.8	275
088742	20	77	4.8	355
088743	223	28	5.9	91
088744	106	49	9.5	845
088745	80	67	5.6	615
088746	29	29	.1	6
088747	57	80	1.2	82
088748	100	167	11.9	1180
088749	84	37	3.5	455
088750	38	10	.7	83
088751	29	7	.1	9
088752	15	2	.1	1
088753	19	3	.1	3
088754	3	8	.1	1
088755	3	7	.1	1
088756	7	5	1.1	185
088757	3	8	.1	27
088758	5	5	.1	32
088759	5	7	3.8	650
088760	7	4	3.3	475
088761	6	6	4.8	830
088762	19	5	4.8	625
088763	19	9	3.4	510
088764	97	20	.7	128
STD C/AU-R	60	41	6.8	485

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: FEB 11 1987

DATE REPORT MAILED: Feb 13/87

GEOCHEMICAL ICP ANALYSIS

.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, CA, P, CR, MG, BA, TI, B, AL, NA, K, W, SI, ZR, CE, SN, Y, NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: CORE AU* ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Speyer* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON PROJECT-SADIM FILE# 87-0290

PAGE 1

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
088765	158	24	.3	43
088766	69	45	.9	14
088767	69	38	.4	60
088768	13	15	.1	16
088769	11	12	.2	20
088770	53	32	.9	21
088771	67	12	.4	24
088772	57	3	.1	1
088773	40	5	.1	7
088774	67	4	.1	6
088775	88	5	.1	1
088776	103	5	.1	1
088777	80	5	.1	2
088778	96	8	.1	7
088779	104	5	.1	25
088780	75	7	.1	7
088781	100	13	.1	1
088782	102	2	.1	1
088783	118	8	.1	2
088784	80	10	.1	1
088785	73	7	.1	13
088786	100	12	.1	1
088787	8	7	.1	1
088788	74	2	.1	1
088789	47	2	.1	1
088790	46	4	.1	1
088791	105	9	.6	82
088792	83	5	.3	97
088793	86	4	17.3	99
088794	50	4	.1	1
088795	100	3	.1	1
088796	96	5	.1	1
088797	75	5	.1	2
088798	67	2	.1	1
088799	150	1526	37.1	3620
088800	59	2	.1	40
STD C/AU-R	59	38	7.0	490

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
088951	73	8	.3	1
088952	20	6	.2	1
088953	65	2	.3	1
088954	98	8	1.1	87
088955	48	4	.3	1
088956	30	10	.2	1
088957	78	2	.2	1
088958	52	6	.4	65
088959	19	5	.1	1
088960	445	8	1.3	155
088961	231	4	1.0	106
088962	74	6	.4	92
088963	76	2	.1	19
088964	102	4	.1	1
088965	77	10	2.6	350
088966	120	5	1.3	155
088967	54	6	.1	17
088968	126	8	1.4	195
088969	92	12	1.9	245
088970	46	7	.5	76
088971	119	6	.1	41
088972	172	40	5.5	465
088973	401	25	9.1	675
088974	116	2977	31.8	3690
088975	376	124	8.1	665
088976	259	83	6.8	108
088977	103	16	3.3	156
088978	69	24	1.0	49
088979	16	29	.1	5
STD C/AU-R	60	40	7.0	515

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: FEB 12 1987

DATE REPORT MAILED: Feb 17/87...

GEOCHEMICAL ICP ANALYSIS

.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, CA, P, CR, MG, BA, TI, B, AL, NA, K, W, SI, ZR, CE, SN, Y, NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: CORE AU* ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. Toy* DEAN TOYE. CERTIFIED B.C. ASSAYER.

I.M. WATSON PROJECT - SADIM 87 FILE # 87-0323 PAGE 1

SAMPLE#	Cu PPM	Pb PPM	Zn PPM	Au* PPB
088851	98	6	74	2
088852	75	6	67	3
088853	83	10	59	10
088854	110	8	52	2
088855	109	9	56	4
088856	82	9	52	5
088857	232	7	63	1
088858	139	7	64	1
088859	93	9	71	1
088860	107	3	75	1
088861	102	3	76	2
088862	101	5	71	1
088863	115	4	76	3
088864	138	4	78	1
088865	62	2	83	1
088866	85	5	66	2
088867	84	6	62	2
088868	103	4	77	1
088869	86	6	71	1
088870	130	7	57	26
088871	94	4	57	15
088872	89	4	68	2
088873	92	8	70	1
088874	67	6	77	1
088875	65	10	39	61
088876	57	7	57	51
088877	54	5	71	7
088878	21	7	60	5
088879	146	8	82	2
088880	82	4	65	52
088881	80	5	46	98
088882	19	5	40	1
088883	14	2	30	1
088884	8	6	32	2
088885	61	2	30	1
088886	2	4	27	1
STD C/AU-R	60	39	136	520

SAMPLE#	Cu PPM	Pb PPM	Zn PPM	Au* PPB
088887	3	8	33	1
088888	2	10	23	3
088889	1	3	26	1
088890	3	11	42	1
088891	49	8	32	23
088892	20	10	25	2
088893	6	8	40	3
088894	7	11	30	2
088895	7	11	45	12
088896	3	12	42	1
088897	3	6	45	2
088898	4	12	44	1
088899	595	11	73	54
088900	54	4	81	28
088995	79	5	85	7
088996	94	6	89	3
088997	76	6	74	1
088998	74	6	71	3
088999	87	7	67	33
089000	76	25	285	4
088980	25	13	49	2
088981	46	7	57	1
088982	51	7	57	13
088983	60	9	58	11
088984	45	7	46	9
088985	39	11	46	6
088986	21	13	27	20
088987	53	15	72	24
088988	122	14	83	10
088989	135	19	80	3
088990	195	2	63	1
088991	95	86	76	1
088992	78	4	66	1
088993	76	2	55	1
088994	50	2	61	1
STD C/AU-R	61	37	137	505

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE 253-3158 DATA LINE 251-1011

DATE RECEIVED: FEB 16 1987

DATE REPORT MAILED: Feb 18/87

GEOCHEMICAL ICP ANALYSIS

.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.CA.P.CR.MG.BA.TI.B.AL.NA.K.W.SI.ZR.CE.SN.Y.NB AND TA. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: CORE AU* ANALYSIS BY AA FROM 10 GRAM SAMPLE.

ASSAYER: *D. J. ...* DEAN TOYE. CERTIFIED B.C. ASSAYER.

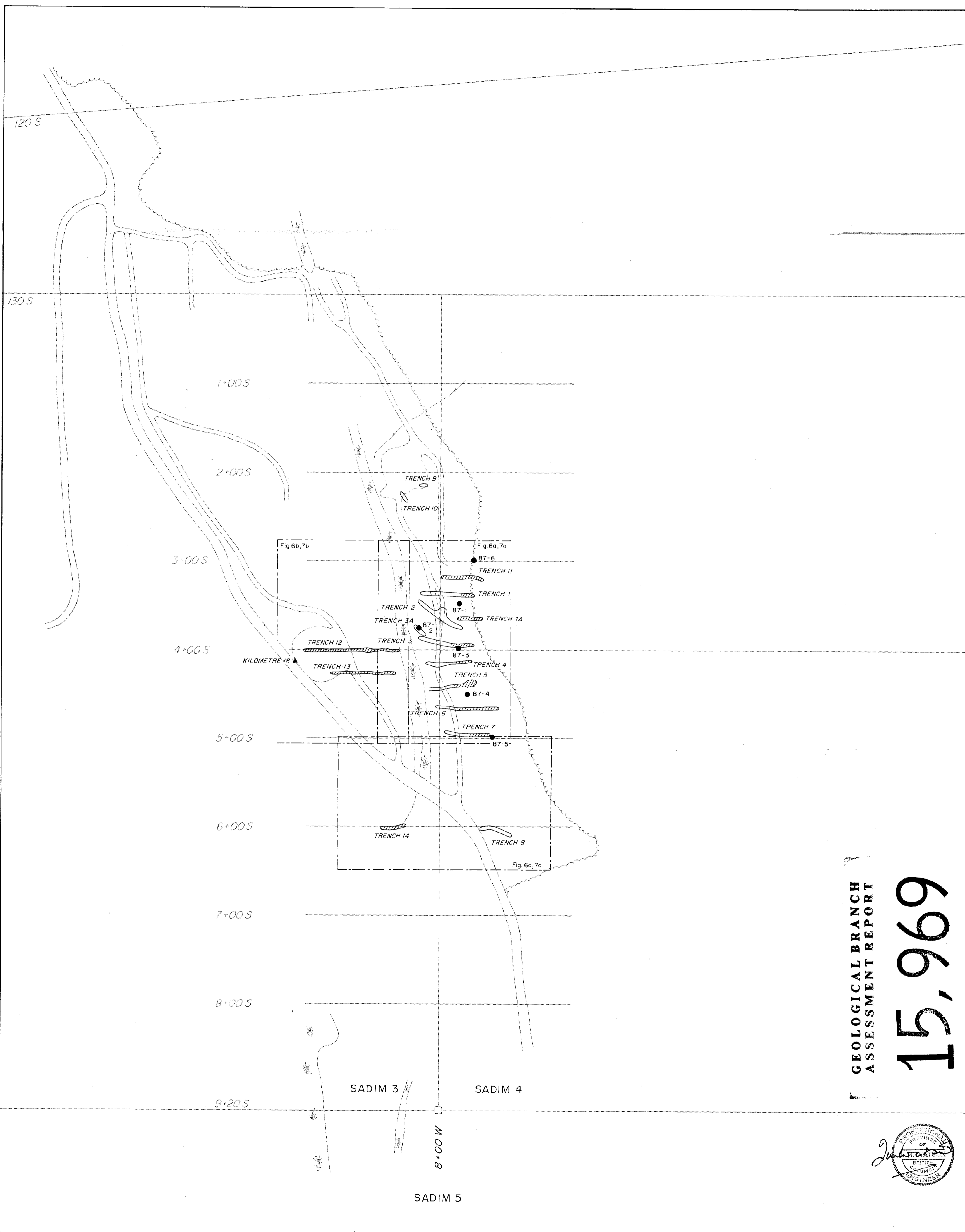
I. M. WATSON

PROJECT - SADIM FILE # 87-0350

PAGE 1

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
091101	4	9	.2	7
091102	49	29	.4	30
091103	287	24	.8	50
091104	153	10	.5	24
091105	108	11	.4	58
091106	71	13	.3	15
091107	157	98	2.0	19
091108	53	21	.5	18
091109	49	11	.5	26
091110	90	55	2.4	265
091111	93	13	1.1	104
091112	119	7	.8	124
091113	48	6	.2	32
091114	44	6	.1	37
091115	60	9	.7	44
091116	71	6	.3	31
091117	61	12	2.1	250
091118	10	6	.1	2
091119	2	11	.1	1
091120	27	16	.7	85
091121	3	12	.1	4
091122	63	9	1.7	235
091123	87	4	1.7	265
091124	112	8	1.1	125
091125	63	11	1.6	215
091126	71	12	1.2	165
091127	39	25	1.6	225
091128	20	15	1.5	185
091129	35	13	.7	98
091130	50	16	8.7	1310
091131	7	11	.1	7
091132	8	11	.1	1
091133	27	13	.3	42
091134	43	11	.7	67
091135	14	12	.1	3
091136	15	10	.8	112
STD C/AU-R	57	41	6.9	495

SAMPLE#	Cu PPM	Pb PPM	Ag PPM	Au* PPB
091137	56	14	3.7	560
091138	89	53	1.0	76
091139	27	37	.5	7
091140	52	37	1.0	129
091141	49	28	1.2	127
091142	37	41	1.2	77
091143	75	62	4.5	580
091144	81	11	.1	32
091145	45	6	.2	1
091146	43	6	.1	1
091147	31	3	.1	3
091148	27	6	.1	2
091149	54	3	.2	1
091150	75	11	.3	1
091151	19	11	.1	1
091152	29	5	.2	1
091153	63	7	.1	1
091154	104	7	.2	1
091155	47	5	.1	1
091156	25	8	.1	10
091157	15	13	1.6	285
091158	29	7	.1	19
091159	20	10	.8	120
091160	13	7	.3	46
091161	8	8	.1	2
091162	328	1064	15.0	1630
091163	41	27	3.7	505
091164	101	47	8.0	965
091165	295	298	22.3	2530
091166	169	2552	159.1	19800
091167	62	39	12.9	1750
091168	211	3	3.4	505
091169	48	7	2.0	445
091170	50	403	28.4	3960
STD C/AU-R	59	36	6.9	500



**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

15,969

- Trench - first phase
- Trench - second phase
- 87-2 Diamond drill hole collar
- Map boundaries, 1:200 scale map sheets (trench geology & geochemistry)
- Roads
- Creek
- Swamp
- Edge of clearing

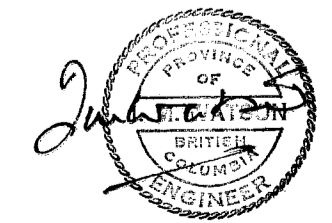
LARAMIDE RESOURCES LTD.
SADIM 3 & 4 CLAIMS SIMILKAMEEN M.D., B.C.

DRILL-HOLE & TRENCHING PLAN

SCALE	DATE	BY	N.T.S.	FIG. N ^o
1:2500	Oct/86 Mar/87	dip IMW	92H/10E	4

Scale 0 100 200 metres

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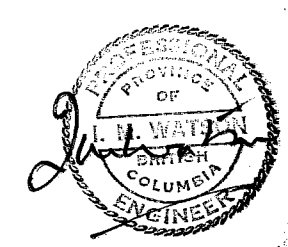


- 1a Green-grey, green, fine- to medium-grained pyroxene andesite
- 1d Green andesitic breccia
- 1df Andesite-limestone breccia
- 1e Green-grey, green andesitic tuffs, often bedded
- 1eal Pale grey-buff calcareous
- 1esil Pale grey, rusty-weathering, fine-grained, silicified and pyritised
- 1f Pale-grey, fine-grained, massive to finely-bedded limestone
- 1g Dark grey, finely-bedded argillites, in part possibly tuffaceous
- 5 Grey to grey-green, fine- to medium-grained pyroxene diorite

- py Pyrite
- ccp Chalcopyrite
- mal Malachite
- gal Galena
- ep Epidote
- Qv/60 Quartz veins, showing attitude
- x Float
- Quartz float
- Outcrop
- 50° Bedding: inclined, vertical
- Joints (vertical)
- Foliation (vertical)
- Trench
- 87-2 Diamond drill hole collar

- ▬▬▬ Roads
- ~~~ Creek
- ▬▬▬ Swamp
- ▬▬▬ Edge of clearing

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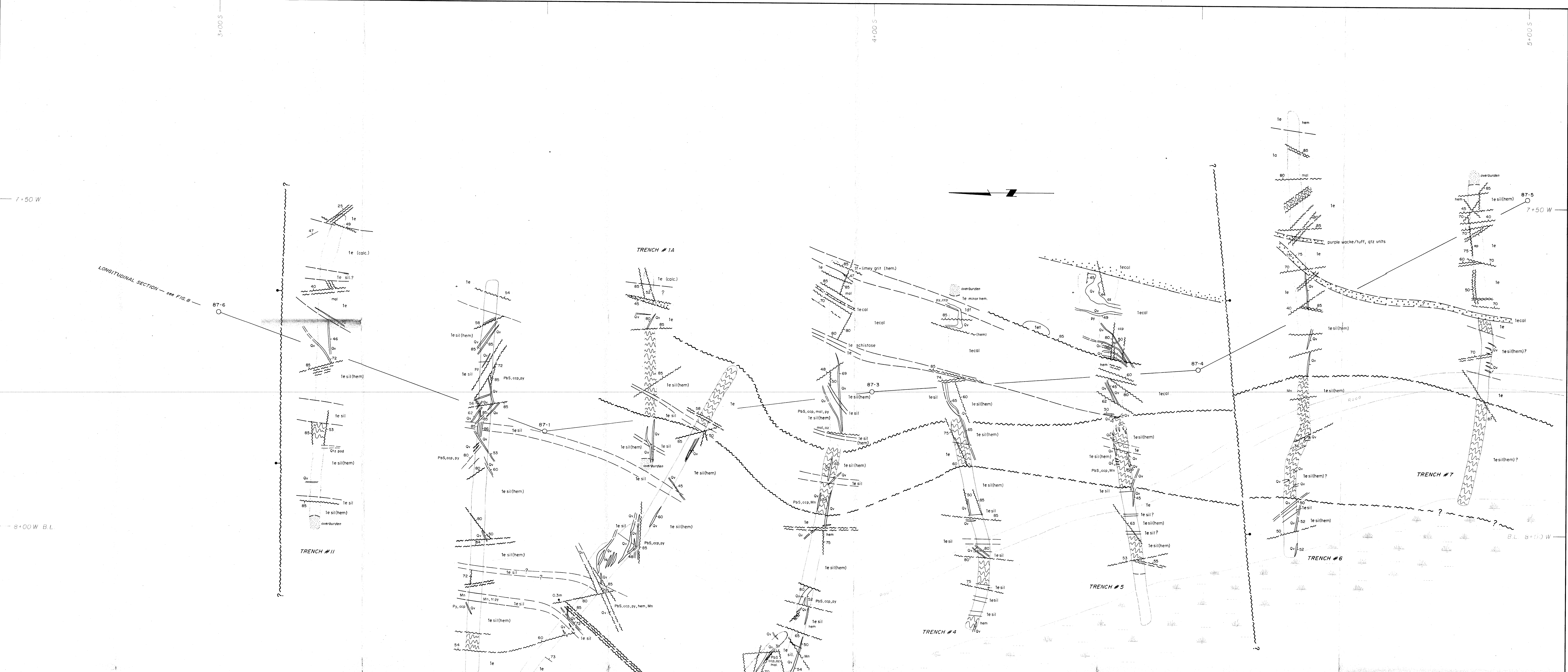
SADIM 3 & 4 CLAIMS, B.C.

GEOLOGY

SCALE	DATE	BY	N.T.S.	FIG. NO
1:2500	Oct/86 Mar/87	dip IMW	92H/IOE	5

Scale 0 100 200 metres

I.M. WATSON & ASSOCIATES LTD.



LEGEND

1a	Green-grey, green, fine- to medium-grained pyroxene andesite	hem	Haematite
1d	Green andesitic breccia	az	Azurite
1af	Andesite - limestone breccia	Mn	Manganese
1e	Green-grey, purple andesitic tuffs, often bedded	py	Pyrite
1ecal	Pale grey - buff calcareous grit - conglomerate bands	cca	Chalcocyanite
1esil	Pale grey, rusty-weathering, fine-grained, silified and pyritised	md	Malachite
1f	Pale grey, fine-grained, massive to finely-bedded limestone	PbS	Galena
1g	Dark grey, finely bedded argillites, in part possibly sulfuraceous	ep	Epidote
5	Grey to grey-green, fine- to medium-grained pyroxene diorite	Q _v /160	Quartz veins; showing attitude

	Bedding: inclined, vertical		Road
	Shear zone		Swamp
	87-2 Diamond drill hole callar		
	Thrust fault, showing downthrow side		

MAP LOCATION KEY

LARAMIDE RESOURCES LTD.
 SADIM 3 & 4 CLAIMS, B.C.
SADIM 3 & 4 EAST
 TRENCHES 1, 1A, 2, 3, 3A, 4, 5, 6, 7, 11
GEOLOGY

SCALE	DATE	BY	N.T.S.	FIG. NO.
1:200	Oct/86 Mar/87	dip IMW	92H/10E	6d

Scale 5 0 5 10 metres
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8+50 W

8+50W

9+00W

9+50W

3+00S

3+00S

4+00S

4+00S

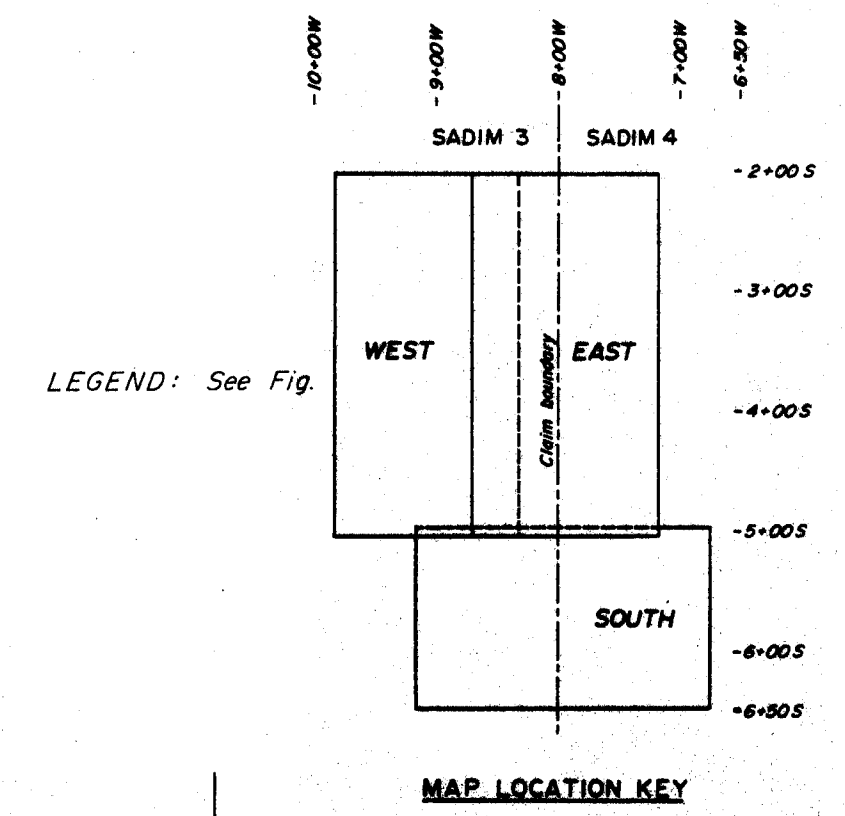
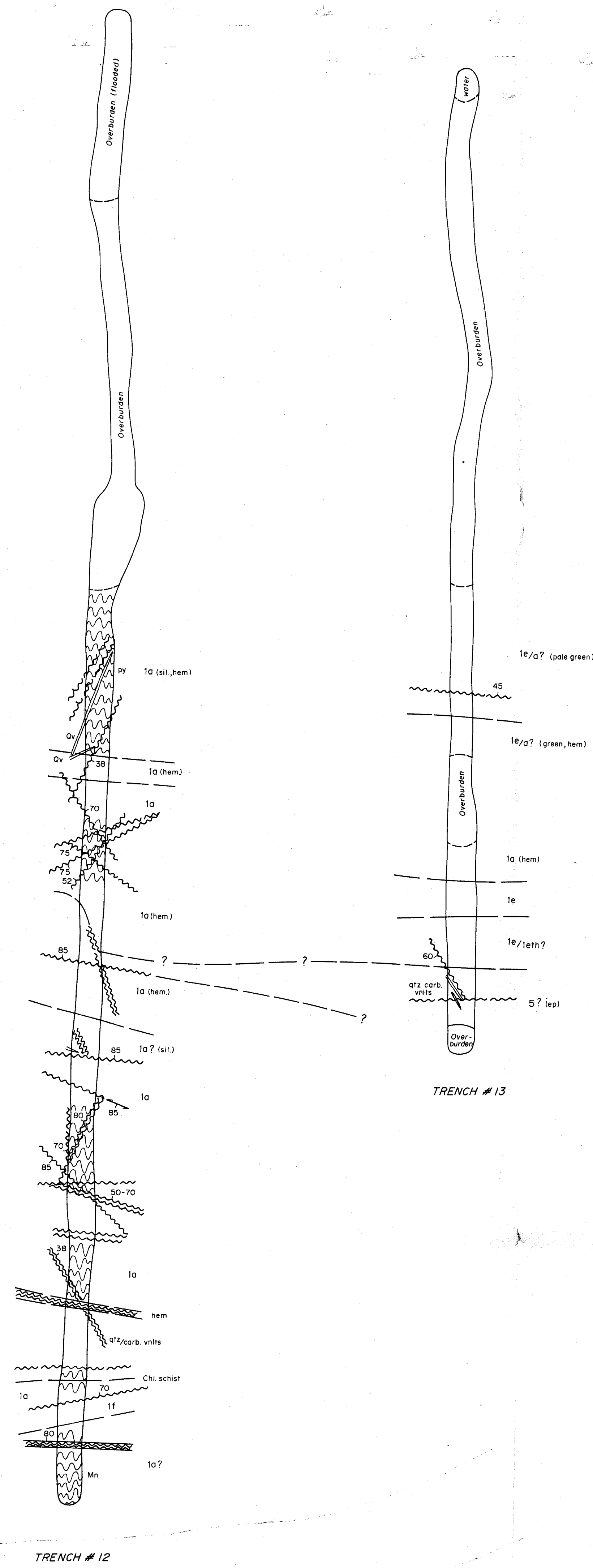
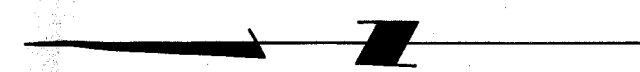
5+00S

8+50W

9+00W

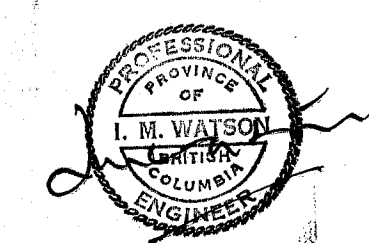
9+50W

5+00S



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Legend: see Fig. 6a

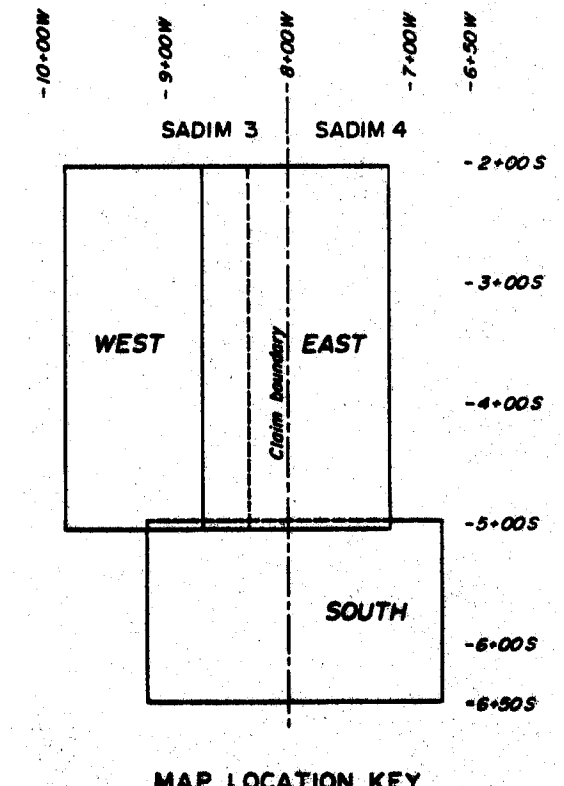
LARAMIDE RESOURCES LTD.				
SADIM 3 & 4 CLAIMS, B.C.				
SADIM 3 & 4 WEST				
TRENCHES 12 & 13				
GEOLOGY				
SCALE	DATE	BY	NTS.	FIG. NO.
1:200	DEC/86 MAR/87	98 IMW	92H/10E	6b
Scale 0 5 10 metres				
I.M. WATSON & ASSOCIATES LTD.				



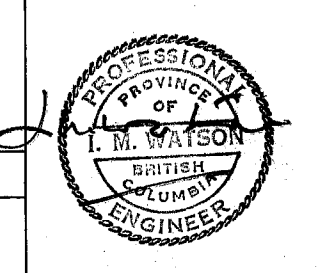
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Legend: see Fig 6a



LARAMIDE RESOURCES LTD.				
SADIM 3 & 4 CLAIMS, B.C.				
SADIM 3 & 4 SOUTH TRENCHES 8 & 14 GEOLOGY				
SCALE	DATE	BY	N.T.S.	FIG. NO.
1:200	DEC/86 MAY/87	dip IMW	92H/IOE	6C
Scale 0 5 10 metres				
I.M. WATSON & ASSOCIATES LTD.				



3+00 S

4+00 S

5+00 S

7+50 W

7+50 W

8+00 B.L.

B.L. 8+00 W

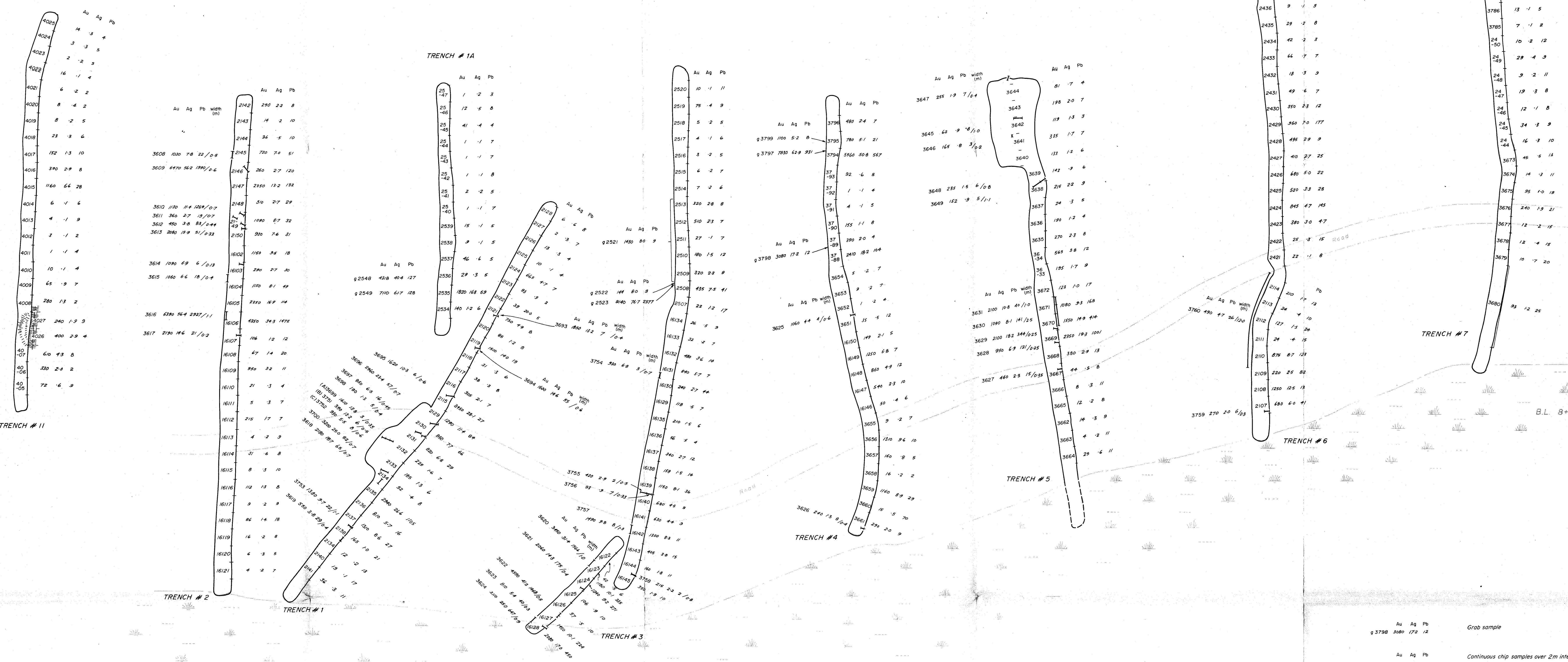
8+50 W

8+50 W

3+00 S

4+00 S

5+00 S

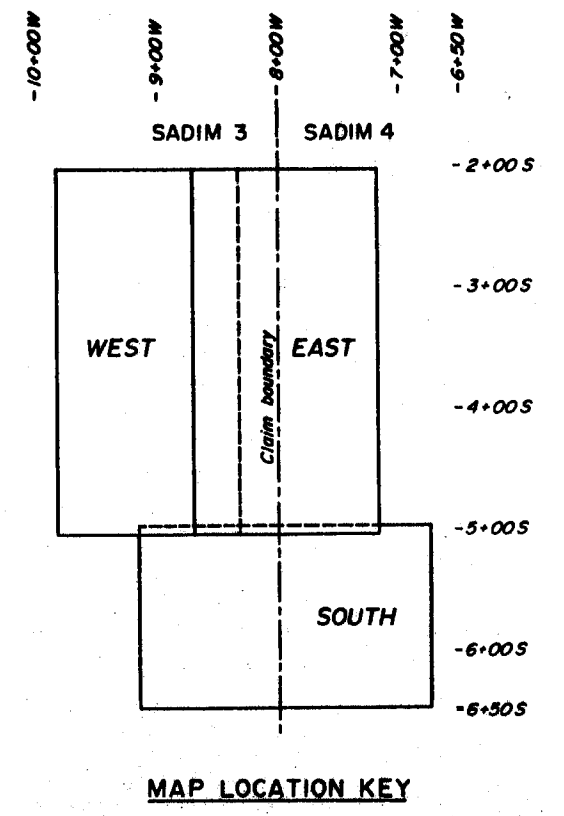


Au Ag Pb 3798 380 172 12	Grab sample
Au Ag Pb 3673 99 5 13	Continuous chip samples over 2m intervals
Au Ag Pb width (cm) 3625 360 4.4 3.4	Vein channel samples

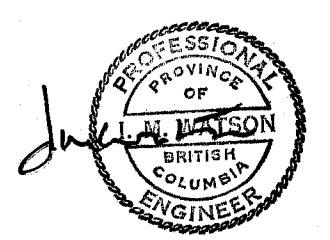
(Au in ppm; Ag, Pb in gpm)

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LARAMIDE RESOURCES LTD.				
SADIM 3 & 4 CLAIMS, B.C.				
SADIM 3 & 4 EAST				
TRENCHES 1, 1A, 2, 3, 3A, 4, 5, 6, 7, 11				
GEOCHEMISTRY-ROCK SAMPLING				
SCALE	DATE	BY	N.T.S.	FIG. NO.
1:200	Oct '86	dip		
	Mar '87	IMW	92H/10E	7a
Scale 1" = 50 metres				
I.M. WATSON & ASSOCIATES LTD.				



8+50W

9+00W

9+50W

3+00S

3+00S

4+00S

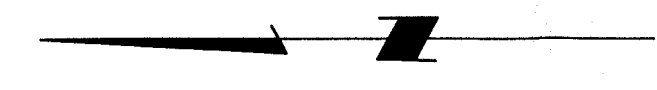
4+00S

5+00S

8+50W

9+00W

5+00S

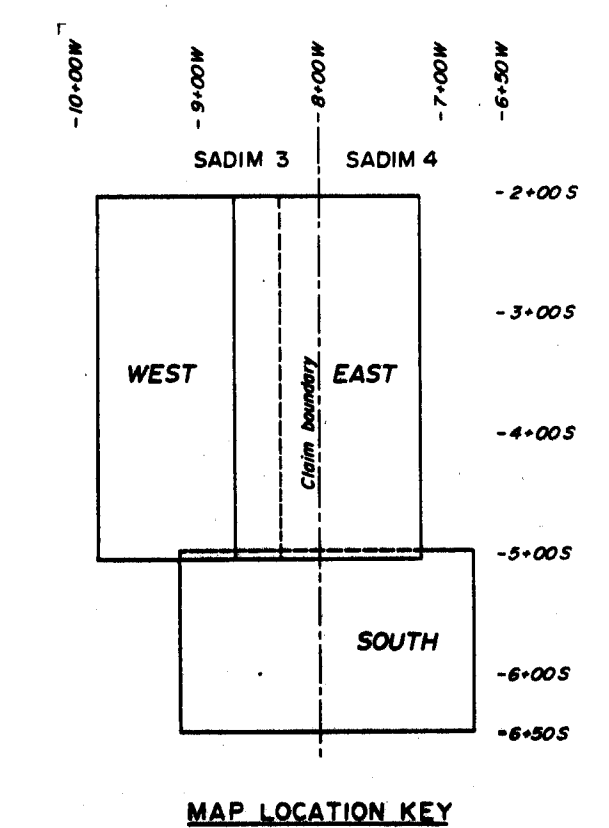


	Au	Ag	Pb
4048 (Qv)	27	✓	✓
40			
4028	21	5	8
4029	13	5	5
4030	19	4	6
4031	8	5	5
4032	60	4	3
4033	71	5	5
40			
4034	2	✓	3
4035	26	✓	2
4036	1	2	3
4037	5	2	2
4038	1	✓	2
4039	12	✓	4
4040	1	✓	2
4041	8	2	3
4042	1	✓	3
4043	3	✓	3
4044	2	3	3
4045	7	✓	5
4046	1	2	5
40			
4047	6	2	5
4048	1	✓	4
4049	1	✓	5
4050	1	✓	5
3761	4	✓	4
37			
3762	6	✓	7
37			
3763	10	✓	6
37			
3764	2	✓	8
37			
3765	1	✓	8
37			
3766	1	✓	5
3767	2	✓	6
3768	2	2	3
3769	5	2	44
3770	3	3	43
3771	2	2	7

TRENCH #12

	Au	Ag	Pb
3779	3	2	4
3780	2	✓	8
3781	18	3	4
3782	1	✓	4
37			
3783	14	✓	4
3784			
3772	7	4	7
3773	43	2	8
3774	6	✓	6
3775	1	✓	4
3776	2	✓	6
3777	1	✓	6
3778	10	✓	10

TRENCH #13

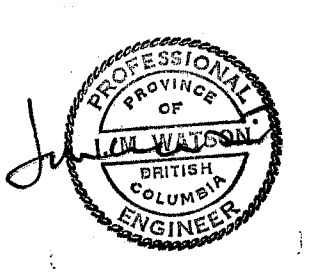


Au Ag Pb
3773 43 2 8
Continuous chip samples over 2m intervals

(Au in ppb; Ag, Pb in ppm)

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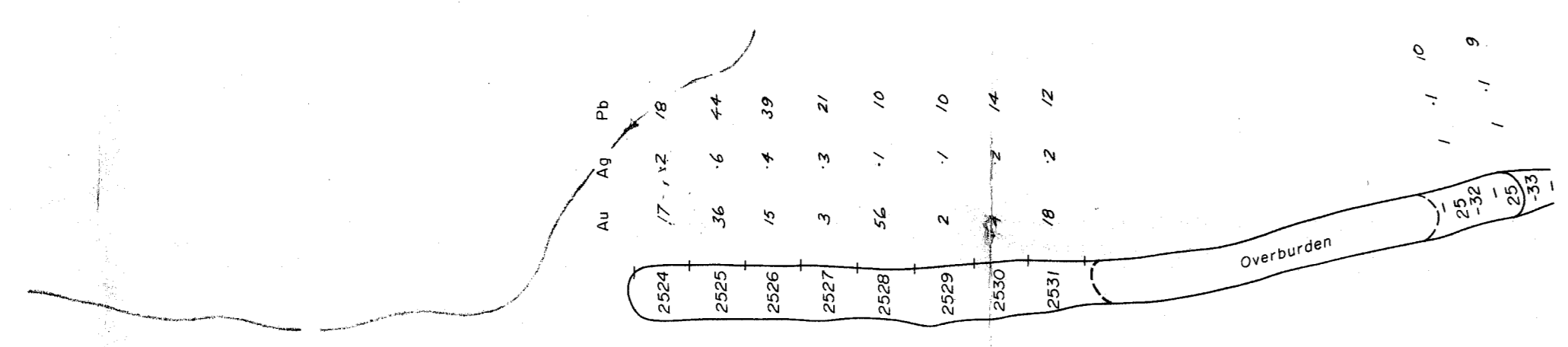
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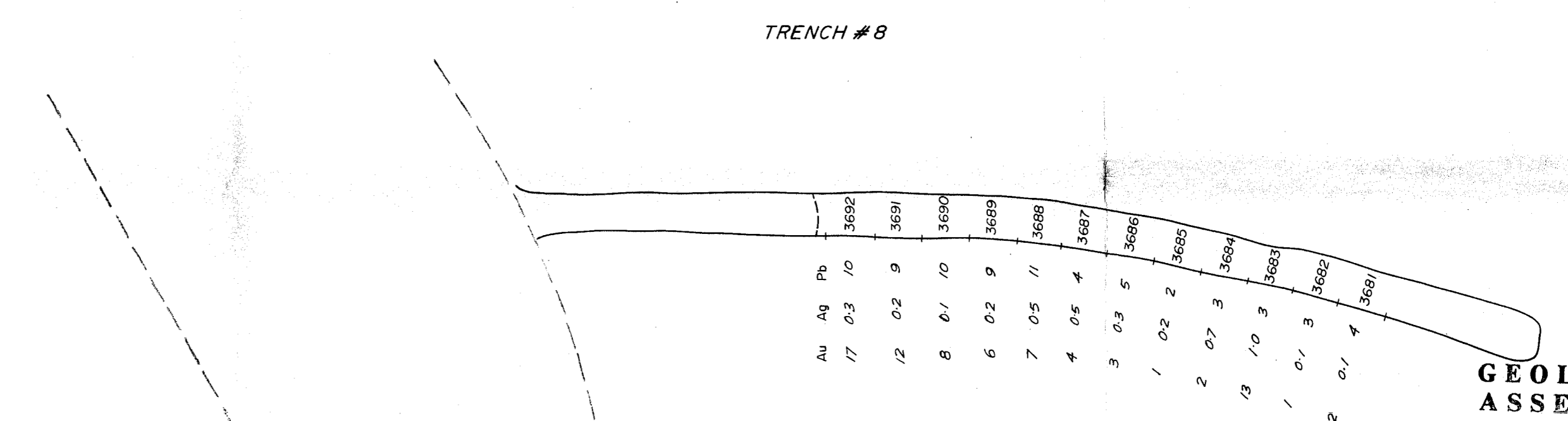
LARAMIDE RESOURCES LTD.				
SADIM 3 & 4 CLAIMS, B.C.				
SADIM 3 & 4 WEST				
TRENCHES 12 & 13				
GEOCHEMISTRY-ROCK SAMPLING				
SCALE	DATE	BY	NTS.	FIG. NO.
1:200	DEC/86	dis	NW	92H/10E 7b
Scale 0 5 10 metres				
I.M. WATSON & ASSOCIATES LTD.				

5+00 S 7+00 W 7+50 W 7+00 W 5+00 S

6+00 S 6+00 S



TRENCH # 14

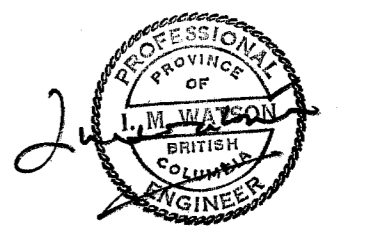
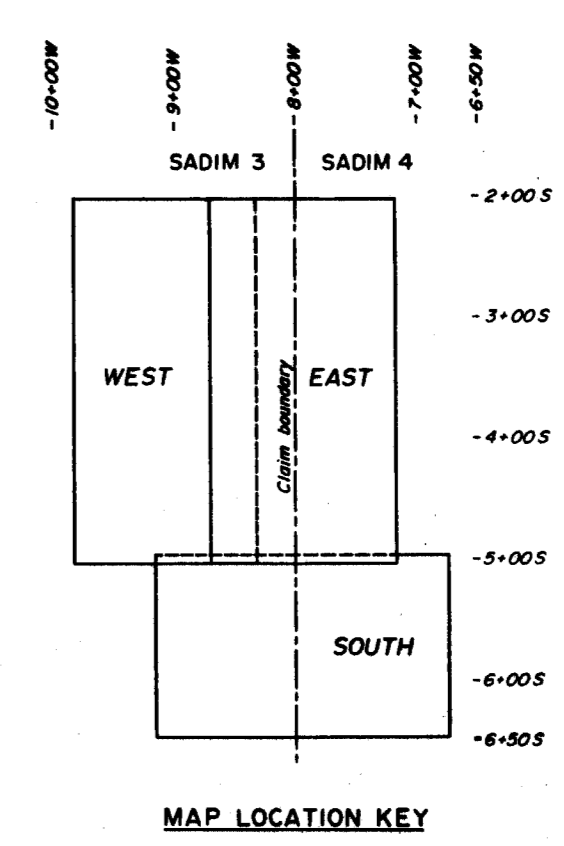


TRENCH # 8

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(Au in ppb; Ag, Pb in ppm)
Continuous chip samples over 2m intervals

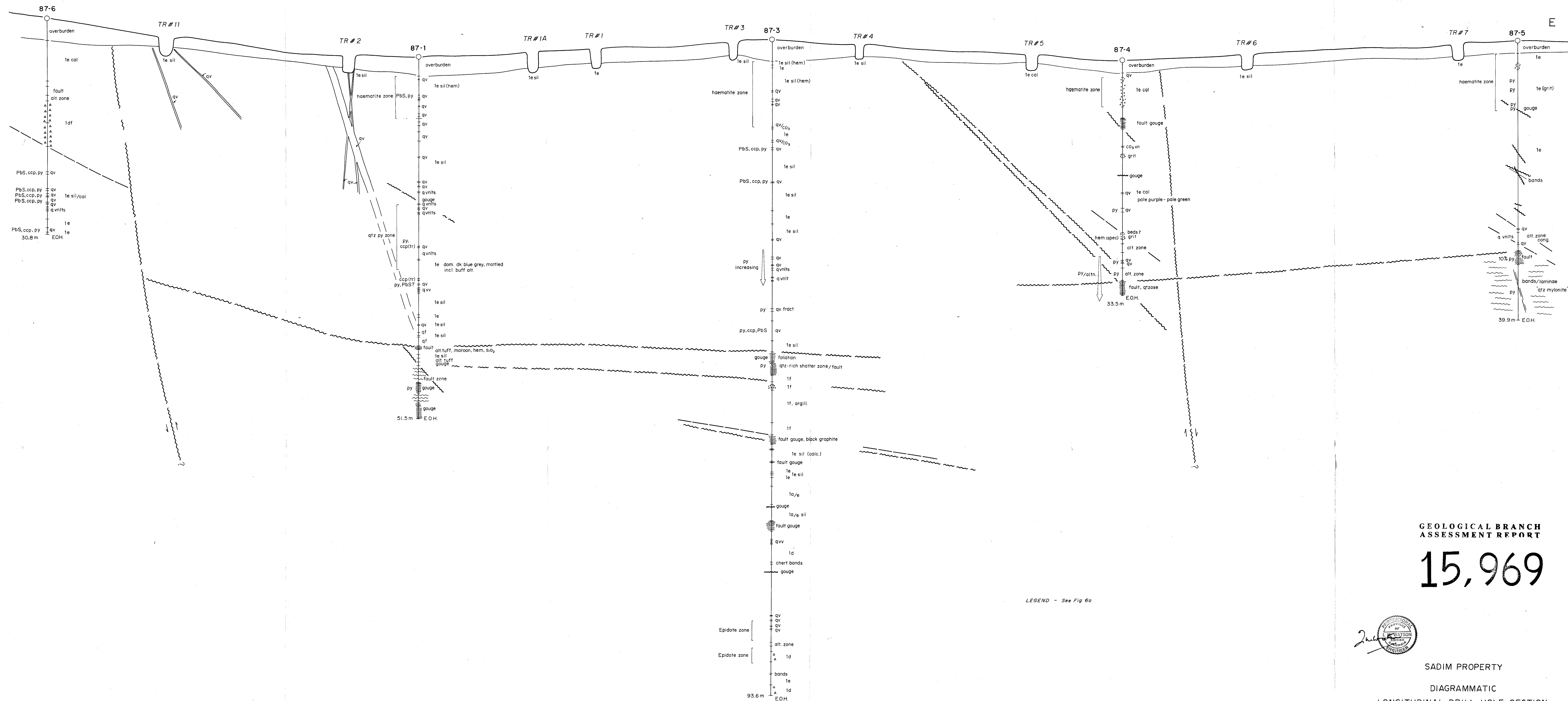


LARAMIDE RESOURCES LTD.			
SADIM 3 B 4 CLAIMS, B.C.			
SADIM 3 B 4 SOUTH TRENCHES 8 B 14			
GEOCHEMISTRY - ROCK SAMPLING			
SCALE	DATE	BY	N.T.S.
1:200	DEC/86 MAR/87	dip IMW	92H/10E
Scale	0 5 10 metres		FIG. NO 7c
LMWATSON & ASSOCIATES LTD.			

M 00+6 M 05+8 M 00+8 M 00+8 7+50 W 7+50 W

W

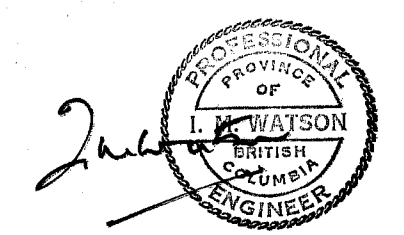
E



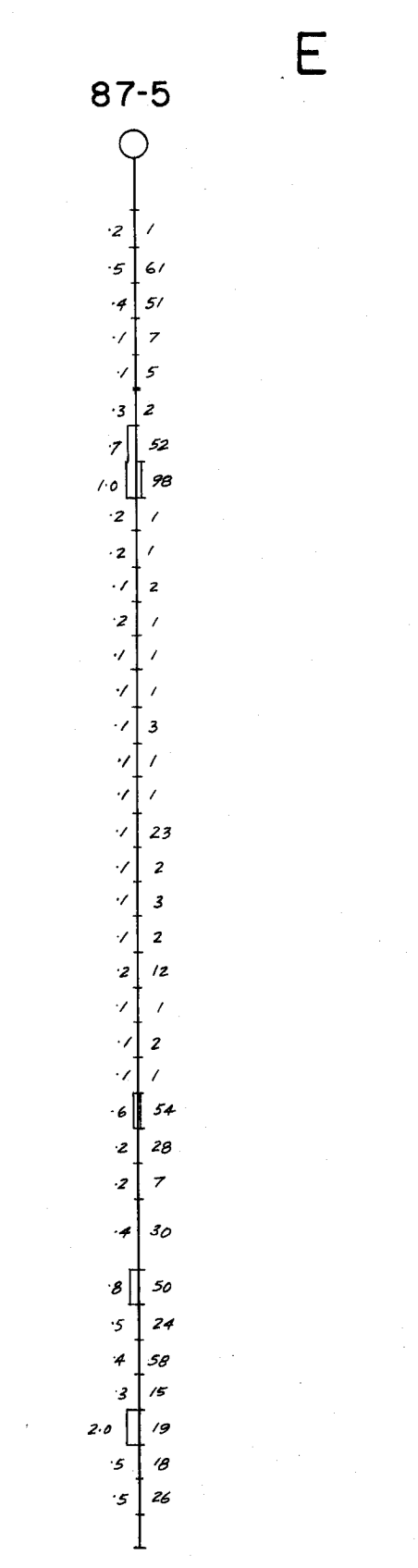
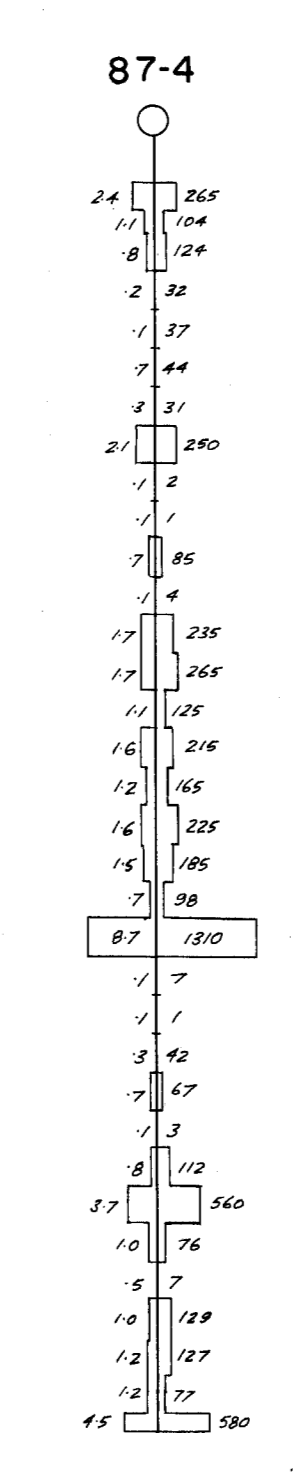
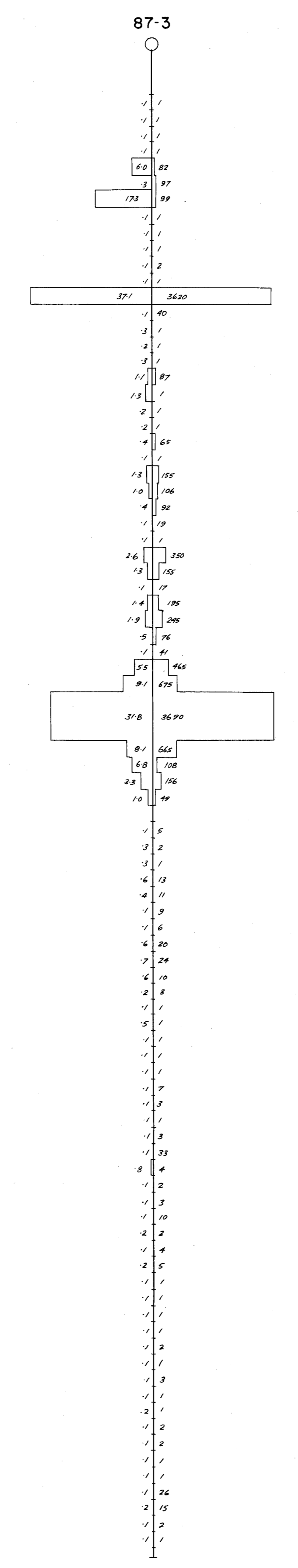
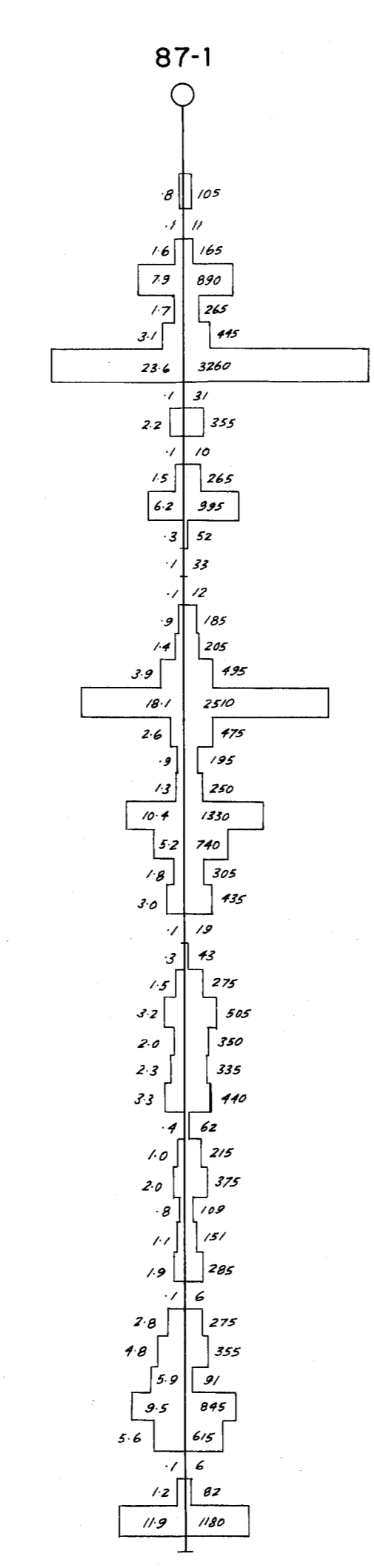
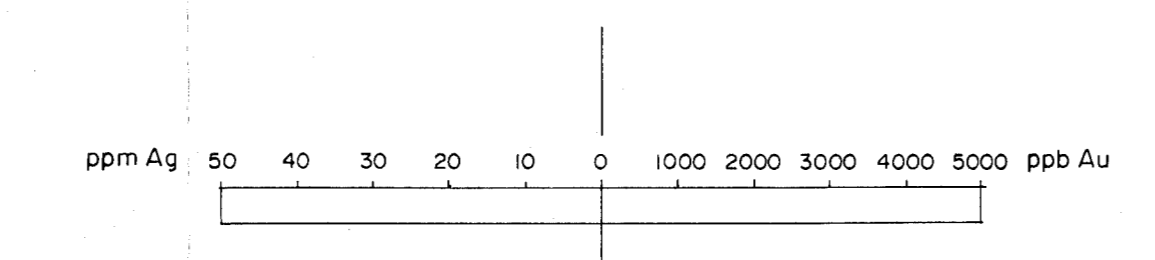
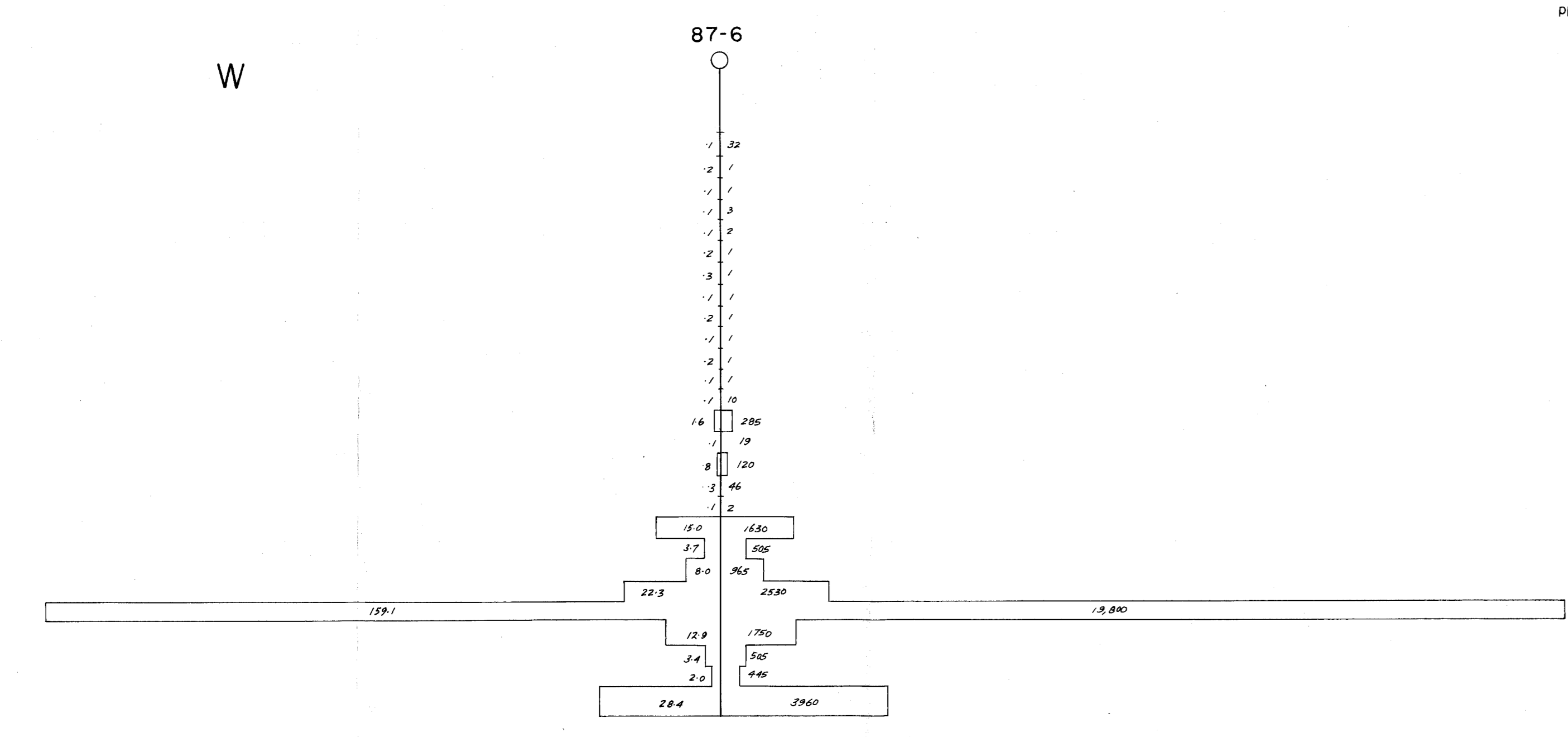
LEGEND - See Fig 6a

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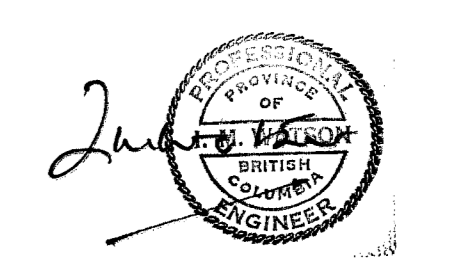


SADIM PROPERTY
DIAGRAMMATIC
LONGITUDINAL DRILL HOLE SECTION
GEOLOGY



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SADIM PROPERTY
DIAGRAMMATIC
LONGITUDINAL DRILL HOLE SECTION
showing Au, Ag assays

Scale 1:200

View North

February 1987

Fig. 8a

W

E

87-6

1d/1d?

overburden

1e calc

Tr.py

Tr.py

fault

alt zone

1df

band'g

1df

1d/n/bdg

carb/qtz vnlt

bx zone

PbS.ccp.py

qv

PbS.ccp.py

qv

PbS.ccp.py

qv

PbS.ccp.py

qv

PbS.ccp.py

qv

30.8 m

EOH

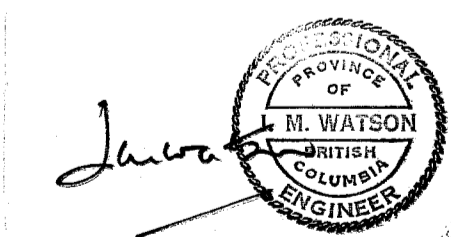
FAULT ZONE?

(projected from Sect'n 3+75 S)

LEGEND - See Fig. 6a

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SADIM PROPERTY

DIAGRAMMATIC CROSS-SECTION: 3+00S

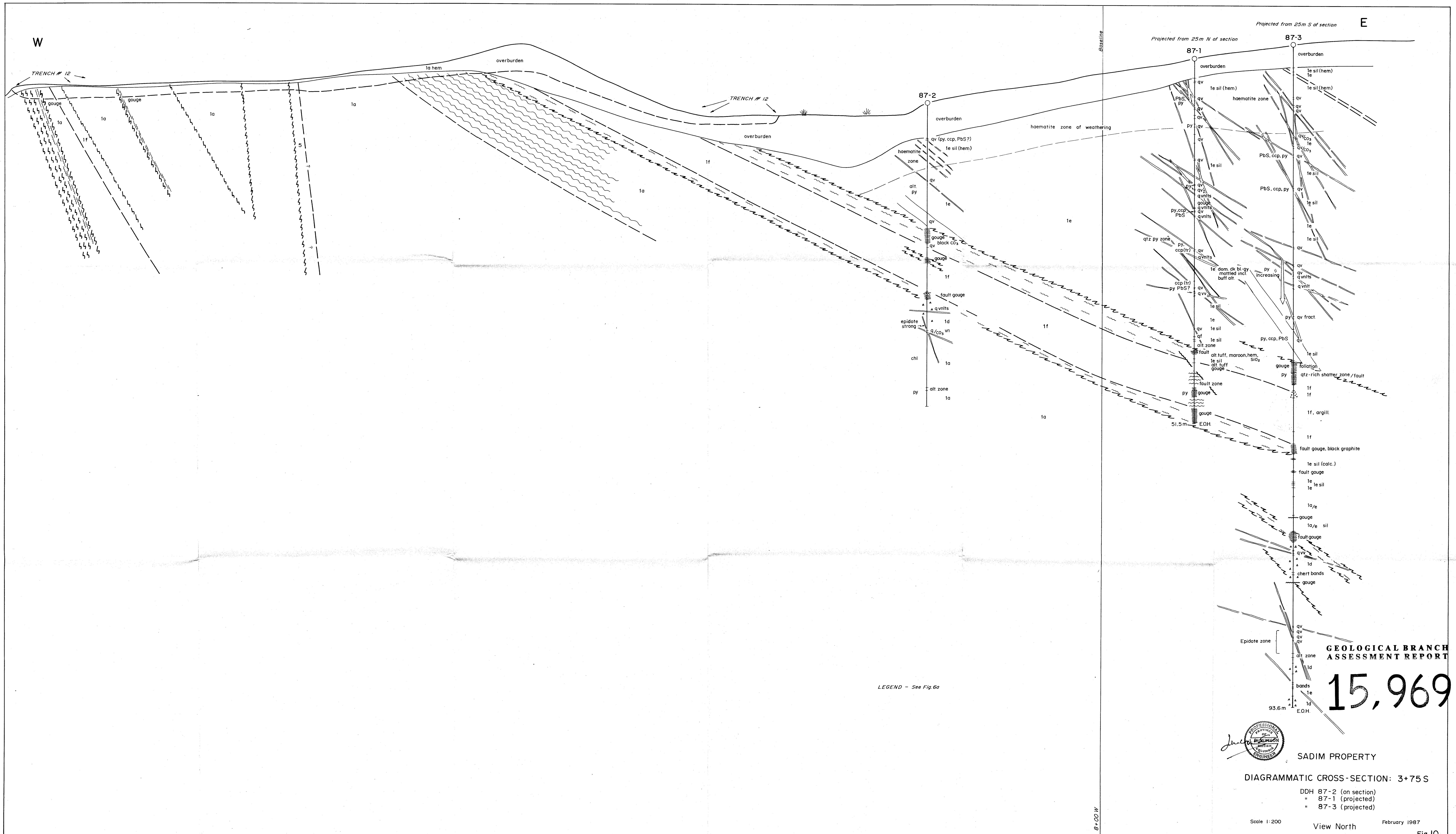
DDH 87-6 (on section)

Scale 1:200

View North

March 1987

Fig. 9



W

E

TRENCH # 12

TRENCH # 12

87-2

87-1

87-3

Projected from 25m N of section

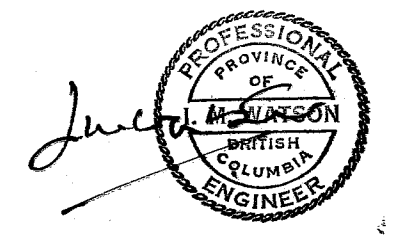
Projected from 25m S of section

Baseline

LEGEND - See Fig. 6a

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SADIM PROPERTY

DIAGRAMMATIC CROSS-SECTION: 3+75 S

DDH 87-2 (on section)
" 87-1 (projected)
" 87-3 (projected)

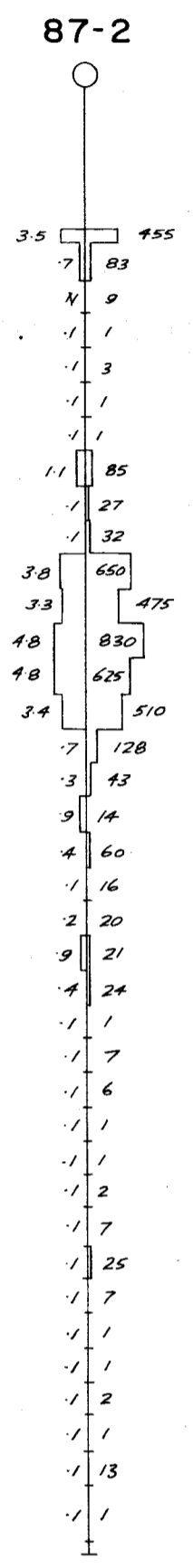
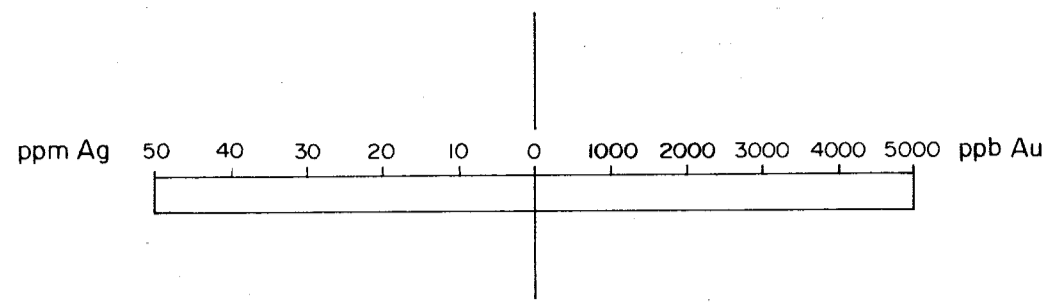
Scale 1:200

View North

February 1987

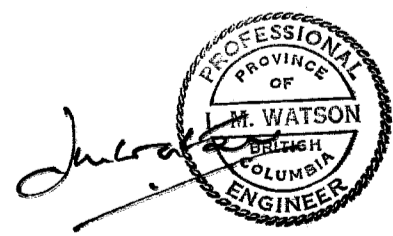
Fig. 10

B+00 W



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SADIM PROPERTY
DIAGRAMMATIC
DRILL HOLE CROSS SECTION 87-2
showing Au, Ag assays