

GEOLOGICAL, GEOCHEMICAL, GEOPHYSICAL AND DIAMOND DRILLING

SUMMARY REPORT

ON THE

MCMASTER ZONE

CAROLIN MINE - LADNER CREEK AREA

NEW WESTMINSTER MINING DIVISION

LAT 49° 32' LONG 121° 17'

CARO #3 FR CLAIM

NTS 92H/11W

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LOG NO: 0409	RD.
ACTION:	
FILE NO:	

For

CAROLIN MINES LTD.

602 - 700 West Pender Street

Vancouver, B.C.

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By

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GEOLOGICAL BRANCH
ASSESSMENT REPORT

19,877

December 6, 1989

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EXECUTIVE SUMMARY

The McMaster Zone was found by soil geochemistry in 1975. Relatively high-grade surface assay values, such as; 20 feet averaging 0.54 oz/ton Au were returned from three bulldozer trenches across the zone. The results of a 1,699 foot (seven holes) diamond drill program, which intersected less intense mineralized sections, was interpreted, at the time, to indicate lack of continuity of the sulfide-gold system. The McMaster Zone was recognized as having very similar gold content, sulfide mineralogy, alteration assemblage, host rocks and stratigraphy as the Idaho Zone. However, no work was done on the McMaster Area between November 1975 and September, 1989.

In 1989, the 1975 McMaster core was relogged and the results correlated with new detailed mapping of the old trenches. This re-interpretation suggested that the McMaster area was composed of several fault wedges separated by westerly dipping shear zones. The easterly dipping mineralized zones exposed in the trenches are truncated by the westerly dipping shears and consequently the 1975 drilling penetrated mineralized zones which are not connected with those immediately up-slope from the drill hole collars.

Diamond drilling of 1,369 feet in six holes was completed to test these new concepts. Five separate mineralized zones were recognized from the surface mapping and labelled A to E. Other zones which are not presently exposed on surface were found in hole M-12, M-13, M-9 and M-11. These mineralized zones would be expected to subcrop west of the trenched area. The 1989 drilling indicated continuity within individual fault wedges and two holes stepped out 76 feet to the north (M-13) and 148 feet to the south (M-12), both of which intersected strong gold values:

M-12	8.50 m to 15.00 m	6.5 m	(21.3 ft.)	averaged 0.152 oz/ton Au
M-13	0.91 m to 3.00 m	2.09 m	(6.9 ft.)	averaged 0.124 oz/ton Au
	3.00 m to 12.50 m	9.5 m	(31.2 ft.)	averaged 0.037 oz/ton Au
	12.50 m to 14.34 m	1.84 m	(6 ft.)	averaged 0.160 oz/ton Au.

The strike length of mineralized zone investigated by the drilling to date is 195 m (640 ft.). Down dip extentions are to a maximum of 40 m (131 ft.) as presently tested by the shallow drilling.

Strong gold-in-soil geochemistry suggest the possibililty of extentions of the mineralized zones of up to 500 ft. to the northwest and at least 600 feet to the southeast.

Current work is not detailed or systematic enough to allow a mineral inventory to be calculated. Considering the possible strike length, number of mineralized zones, width of mineralization, gold grades and known extent down dip and general geological parameters, in my opinion, the McMaster Zone has the potential to contain a mineral deposit similar in size and grade to the Idaho Zone which was developed into the Carolin Mine.

A program of continued geological mapping and surface trenching (with a tracked excavator) in conjunction with additional exploration drilling is recommended for 1990 at a cost of \$353,500 (Canadian). If this program is successful in extending the McMaster Zone along strike and down dip then a major program of definition drilling will be required to block out potential ore zones.

Respectfully submitted,

December 6, 1989

J.T. Shearer, M.Sc., FGAC
New Global Resources Ltd.

INTRODUCTION

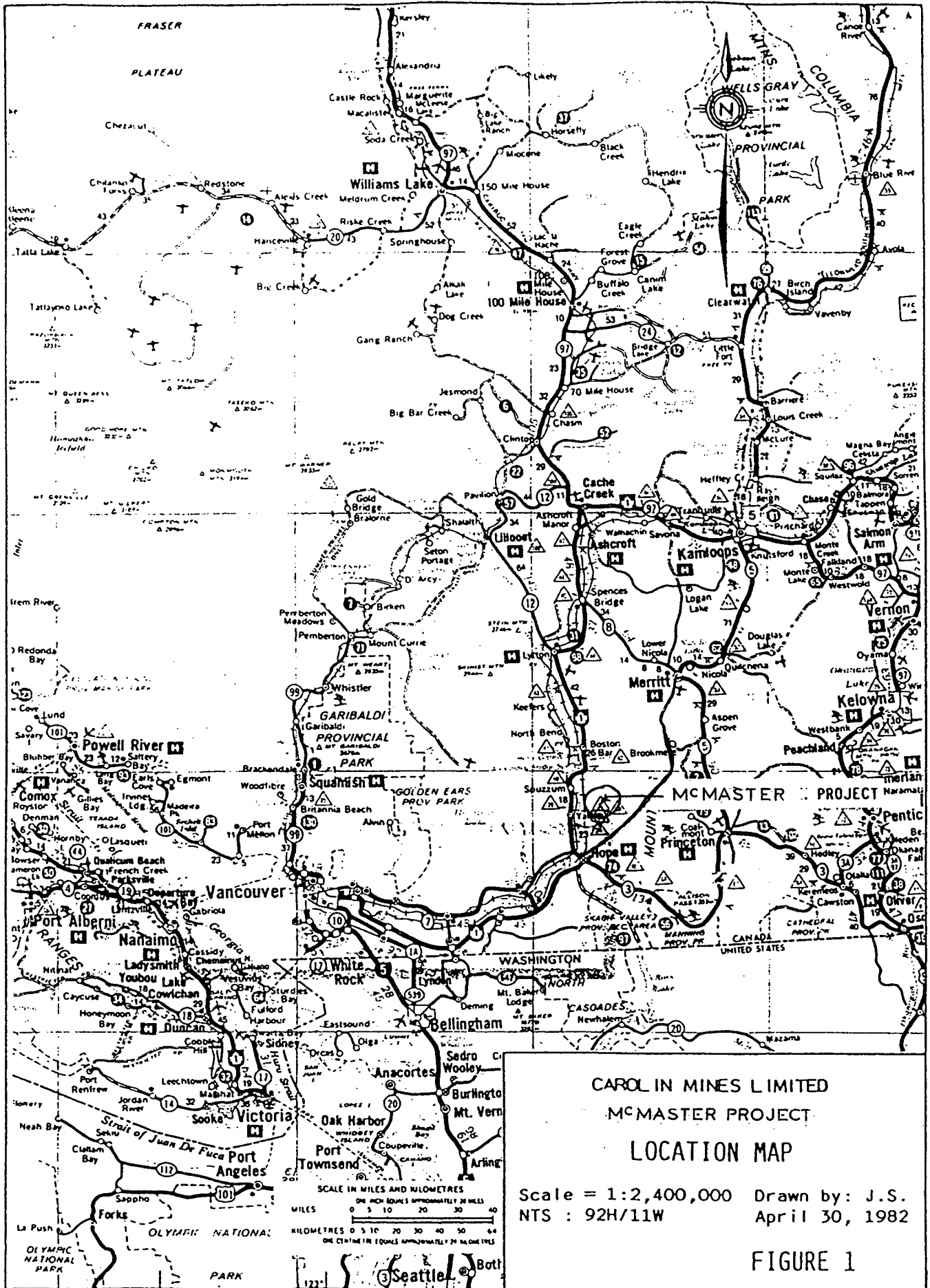
This summary report on the McMaster Zone incorporates a re-interpretation of the 1975 drill data in addition to geological observations, drill logs and complete assay results from the 1989 program.

The McMaster Zone has been known as an important exploration target since its discovery in 1975. Unfortunately, the initial enthusiasm for the zone, created by the relatively high-grade surface trenching results, decreased dramatically after the small diamond drilling program in November 1975 failed to demonstrate continuity of the higher grade sections. The data collected in 1975 did not present a "simple" interpretation comparable to the gross geological simplifications that were being made in the same time period at the Idaho Zone.

The McMaster Zone was discovered by soil geochemistry in early 1975 while the property was under option to Precambrian Shield Resources Ltd. The Zone is located approximately 1,010 meters (3,314 feet) horizontally north of the presently known north end of the Idaho orebody (on 800 level at 900N). The McMaster Zone outcrops at about 1,482 metres ASL (4,862 feet) which places it 431 metres (1,414 ft.) above the Idaho #2 zone outcrop.

The intense gold-in-soil anomaly at the McMaster was investigated by 900 feet of surface excavations in three easterly trending bulldozer trenches. Several discrete quartz-albite-carbonate alteration zones containing abundant pyrite, pyrrhotite and arsenopyrite were uncovered. The style of alteration, sulfide assemblage and host rock sequence are identical to that found in the Idaho orebodies. The trenching demonstrated a strike length of the mineralized zones of approximately 400 feet. Better grade sections in the trenches returned assays ranging from 0.25 oz/ton gold over an 11 foot width to 0.54 oz/ton gold over a 20 foot width.

In November 1975, a total of 1,699 feet of surface diamond drilling in seven holes tested below the trenches to a maximum depth of 300 feet. Similar alteration was encountered in the drill holes but sulfide mineralization was not as intense. The best intersection was 0.13 oz/ton Au over 19.4 feet (M-2) within which 7.1 feet averaged 0.245 oz/ton Au. Unfortunately, since no detail geological mapping was



completed in and around the trenches, the significance of the drill results could not be correlated into an overall picture. Surprisingly, no further work was permitted at the McMaster Zone until 1989.

The McMaster Zone was re-evaluated in September and October, 1989 by the following program:

- a) re-logging all of the 1975 drill core
- b) detail mapping of the trenches at 1:500
- c) detail mapping around the zone 1:1000
- d) plotting cross-section at 1:250 and plans
- e) search of available records for old McMaster data
- f) diamond drilling, 1,369 feet in six holes (M-8 to M-13).

This report includes all past and present information on the McMaster area and recommendations for a staged evaluation of the favourable ore potential of the mineralized zones.

LOCATION AND ACCESS

The Ladner Creek North Property is situated between the headwaters of Ladner Creek to the south and upper reaches of the south fork of Siwash Creek to the north. The McMaster Zone at latitude $49^{\circ} 31' 10''$, longitude $121^{\circ} 17' 45''$ is in the north-central portion of the claim group. The property is 20 km northeast of Hope, B.C., and lies adjacent on the north of the Carolin Mine site as shown on Figures 1 and 2. Elevations in the immediate area range from 1,200 to 1,510 m.

Access from Hope is by the new Coquihalla Highway along the old Kettle Valley Railway grade to km 20 and then up the mine road to the Carolin Mine site. From the mine, a 4-wheel drive gravel and dirt road 8 km long leads north to the McMaster Zone. The west and north sides of the property are accessible by logging roads up Qualark and Siwash Creeks (Figure 3).

CLAIM STATUS

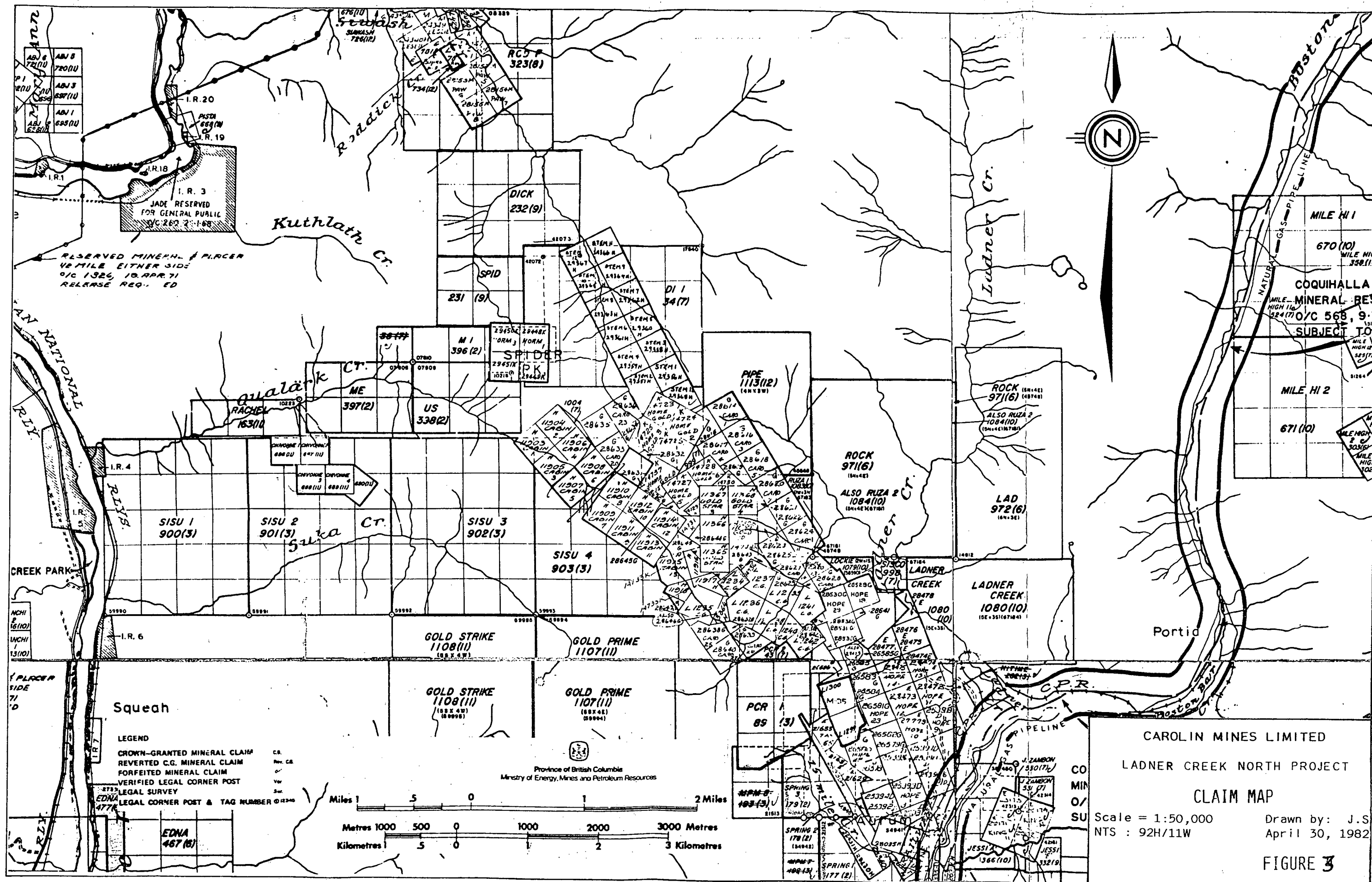
Recently, Carolin Mines Ltd. completed a financial reorganization and asset consolidation, which among other things, enabled the company to arrange 100% ownership in the Carolin mine, mill complex and mineral claims in the Belt. The company also acquired the remaining mineral claim interests in the Ladner Creek North Property and a portion of the Coquihalla Belt property. A partial list of claims is shown in Table 1 and illustrated on Figure 3. The Government-issued claim map is significantly different than the actual claim positions. Carolin Mines Ltd. contracted a legal survey of the common boundary between Ladner Creek North Project and the claims surrounding the Idaho Mine to Tunbridge and Tunbridge Ltd. in 1982. Between June and August most of this boundary, referred to as "the Fence" was located and brushed out and is located just north of McMaster Pond. The relative positions of Cabin 9 and 10, plus Home Gold 5 and 6, were established, but the relatively minor deviation through Caro 17 and Home Gold 15 remain to be calculated. The extreme easterly portion of the Fence along Caro 5 and 6 and a closing loop from east to west have not been surveyed. Fred Tunbridge was involved in staking the Cabin claims with Buster McCombs in 1962.

The Carolin property consists of Crown granted mineral claims, located 2-post claims, Modified Grid System claims and several fractions. The pertinent claim data around the Idaho Mine are as follows:

TABLE 1
LIST OF CLAIMS

<u>Name</u>	<u>Record No.</u>	<u>No. of Units</u>	<u>Expiry Date</u>
	Caro #5 - #16	28618 - 28629	
Caro #24 - #27	28637 - 28640	4	June 29, 1991
Caro #29 & #30	28641 - 28642	2	June 29, 1991
Caro #1 Fr - 5 Fr & 6 Fr	28643 - 28646 - 28647	5	June 29, 1991
Sylvia Fraction	13364	1	July 20, 1991
Cabin #9 - #14	11911 - 11916	6	July 21, 1991
Cabin #20 Fr & #21 Fr	11917 - 11918	2	July 21, 1991
Gold Star No. 1 - No. 4	11365 - 11368	4	July 28, 1991
Home Gold #5 - #14	14727 - 14736	10	August 21, 1991
PCR 1	89	6	March 8, 1991
PCR 2	43	2	July 28, 1991
Idaho, Tramway	1234 - 1235	2	Crown grants
Aurum No. 1 - No. 6	1236 - 1241	6	Crown grants
Monitor	1242	<u>1</u>	Crown grant
	Total	63	

Claims to the north are listed below, Carolin Mines Ltd. also owns claims to the north and south of the listed claims.



LEGEND

CROWN-GRANTED MINERAL CLAIM C.R.

REVERTED C.G. MINERAL CLAIM Rev. CR

FORFEITED MINERAL CLAIM F.

VERIFIED LEGAL CORNER POST Ver.

LEGAL SURVEY S.

LEGAL CORNER POST & TAG NUMBER S.C.

Miles 1 5 0 1 2 Miles

Metres 1000 500 0 1000 2000 3000 Metres

Kilometres 5 0 1 2 3 Kilometres

CAROLIN MINES LIMITED

LADNER CREEK NORTH PROJECT

CLAIM MAP

Scale = 1:50,000

Drawn by: J.S.

April 30, 1982

FIGURE 3

TABLE I

List of Claims

LADNER CREEK NORTH PROJECT

<u>NAME</u>	<u>RECORD NUMBER</u>	<u>UNITS</u>	<u>DATE LOCATED</u>	<u>DATE RECORDED</u>	<u>EXPIRY DATE</u>	<u>LOCATOR</u>
CABIN #1	11903	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #2	11904	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #3	11905	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #4	11906	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #5	11907	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #6	11908	1	July 8, 1962	July 21, 1962	July 21, 1991	A. McCombs
CABIN #7	11909	1	July 12, 1962	July 21, 1962	July 21, 1991	Rae McCombs
CABIN #8	11910	1	July 12, 1962	July 21, 1962	July 21, 1991	Rae McCombs
CARO #1	28614	1	June 13, 1973	June 29, 1973	June 29, 1991	L. McClelland
CARO #2	28615	1	June 13, 1973	June 29, 1973	June 29, 1991	L. McClelland
CARO #3	28616	1	June 13, 1973	June 29, 1973	June 29, 1991	L. McClelland
CARO #4	28617	1	June 13, 1973	June 29, 1973	June 29, 1991	L. McClelland
CARO #17	28630	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #18	28631	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #19	28632	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #20	28633	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #21	28634	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #22	28635	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
CARO #23	28636	1	June 13, 1973	June 29, 1973	June 29, 1991	M. Mathieu
HOME GOLD #1	14723	1	August 6, 1965	August 21, 1965	August 21, 1991	A. McCombs
HOME GOLD #2	14724	1	August 6, 1965	August 21, 1965	August 21, 1991	A. McCombs
HOME GOLD #3	14725	1	August 6, 1965	August 21, 1965	August 21, 1991	A. McCombs
HOME GOLD #4	14726	1	August 6, 1965	August 21, 1965	August 21, 1991	A. McCombs
HOME GOLD #15	14737	1	August 6, 1965	August 21, 1965	August 21, 1991	A. McCombs
DI #1	34	20	July 4, 1975	July 11, 1965	July 11, 1991	D. J. Griffiths
CALEB #1						
FRACTION	999	1	July 10, 1980	July 15, 1980	July 15, 1991	W. F. Chase
CABLE #2						
FRACTION	1004	1	July 18, 1980	July 21, 1980	July 21, 1991	W. F. Chase
STEM #1	29356	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #2	29357	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #3	29358	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #4	29359	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #5	29360	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #6	29361	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #7	29362	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #8	29363	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #9	29364	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #10	29365	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #11	29366	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko
STEM #12	29367	1	June 29, 1974	July 2, 1974	July 2, 1991	G. Beyko

TOTAL 58 units

FIELD PROCEDURES (McMASTER ZONE)

All geological work, diamond drilling and project supervision was done under the author's direct supervision by experienced personnel employed by New Global Resources Ltd.

The 1975 grid could not be accurately reconstituted due to the lack of points still existing in the field. A new grid was established over the area of interest in October, 1989.

A baseline (designed 0+00) was cut along the drill access road paralleling the valley bottom east of the McMaster showing. The baseline trends 135° true north. To the south, the baseline extends beyond the end of the road (Station 18+00N) along the forested hillside. The baseline extends from station L17+00N at its southern extremity to station L20+00N at its northern limit. Crosslines perpendicular to the baseline were cut at 30 metre intervals, between L18+20N and L19+70N. The crosslines extend easterly along azimuth 045° for up to 60 metres and westerly along azimuth 225° to up to 180 metres. Stations were established using pickets at 10 metre intervals along these lines. The lines were measured by a hip-chain.

The 1975 trenches and locations were tied into the grid. Minor slough has covered some of the 1975 drill hole collars making precise locations difficult to determine.

Assay intervals and general locations in the trenches were taken from 1975 Map No. 50F (assay plan) and keyed to mineralized zones. Survey pickets and cairns from the 1975 work were tied-in to the 1989 grid. The 1975 core is stored in the McMaster Camp core shack. Relogging of the 1975 core revealed that many of the major fault structures were not recorded and the four letter rock name abbreviations used by D.J. Griffith did not address the complex lithological assemblage encountered. The new logs reflect the rock nomenclature system built-up during mining of the Idaho ore zones 1981-1984 as established by Shearer (1982) and (1984).

Diamond drill logs are in Appendix VI. The drill contract (Appendix III) was awarded in feet and the core was carefully converted into metric lengths at the Idaho core shack. Core recovery was measured on each piece of core and closely estimated through the uncommon, short rubbly sections.

A drill log form was designed for the project featuring from the left side: drilling blocks, boxes, core recovery, graphic columns for alteration, fracturing, sulfides and geology. The center is reserved for normal written descriptions and assay results are listed on the right. Each drill hole was logged on a scale of 1:250.

Each drill core sample was carefully split by a experienced splitter. New Global personnel checked each sample number with the assay ticket number and each bag was numbered. At the end of the shift all samples were placed in a locked aluminum box. Samples were brought to Chemex Labs by truck. The core shack was locked at all times when New Global personnel were not actually working on the core. Analytical procedures (fire assay) at Chemex Labs Ltd. are outlined in Appendix V. A suite of samples from the McMaster drilling project have been sent to a second independent lab for check assay. The results of the check assays are expected shortly.

REGIONAL GEOLOGY

The Idaho and surrounding claims north to the McMaster Zone cover part of the Coquihalla Serpentine Belt and the early to Middle Jurassic Ladner Group rocks which are adjacent on the east (Cairnes, 1924; Monger, 1970). The two groups of rocks are separated by the Hozameen Fault (Figure 4). This assemblage makes up the main elements of the Coquihalla Gold Belt.

General characteristics of the Ladner Creek area have been discussed by Cochrane and Griffith in numerous Carolin Mines Limited private reports since 1973. Some of these are listed in the bibliography. Surface mapping by Ray (1982, 1983) shows that much of the stratigraphy in the immediate vicinity of the Idaho Orebody is inverted. Major folding and tilting of fault panels appear to be of fundamental importance in ore genesis. A summary of the importance of detail stratigraphic

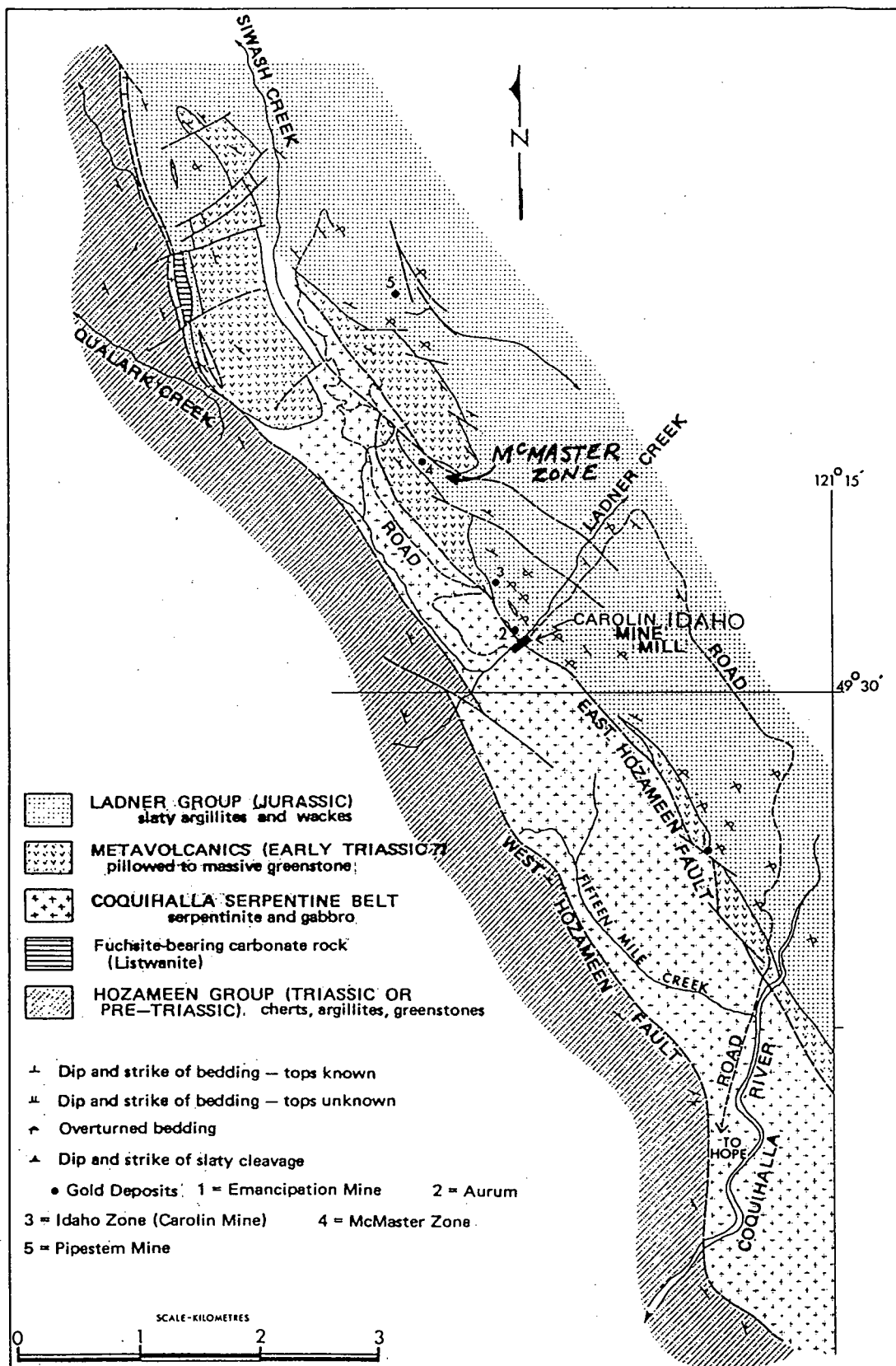


Figure 4 The regional geology of the Carolin-Pipestem-Emancipation gold mines area. (After Ray, 1983).

measurements is contained in Shearer and Niels (1983). The lower Ladner Group rocks represent a transition from a proximal turbidite depositional environment to a progressively distal turbidite and deeper water regime. A regular stratigraphic sequence is recognized within the Ladner Group at the Mine (Figure 8).

The basic structure in the Idaho Mine is a complex, asymmetric antiform which plunges about 20° to the northwest. The ore zones amenable to open longhole stoping are located in the thickened hinge portions of the fold while mineralization generally disappears or thins along the fold limbs (Figure 5). The main fold structure is cut by major late fault structures that run subparallel to the fold axial plane. Cross-cutting faults, trending northeast, appear to be an early element that has moved large blocks of volcanic rocks toward the east.

LOCAL GEOLOGY AND SURFACE MINERALIZATION

Detailed geologic mapping was not done at the McMaster Zone in 1975. The only mapping completed was 1:6,000. Government mapping by G.E. Ray in 1983, Figure 5, was a rapid, one day pace and common traverse. During 1989, systematic 1:500 scale mapping was completed around the trenched area, Figure 9, and a 1:1000 scale general map to the south and west was initiated. A composite map (Figure 7, in pocket) illustrates the relationship between the McMaster area and the Idaho and Pipstem Zones.

It became clear once the 1975 drilling was correlated with the detail trench mapping that the McMaster Zone is a series of thin fault wedges stacked on top of westerly-dipping post-mineralization shear zones. These faults appear to be related to a serpentinite-filled structure which occupies the small valley east of the McMaster mineralized zones. Previous work on the north property suggests in the order of 800 metres of strike-slip movement along this fault (the McMaster Pond Fault). The dip-slip component is presently unknown.

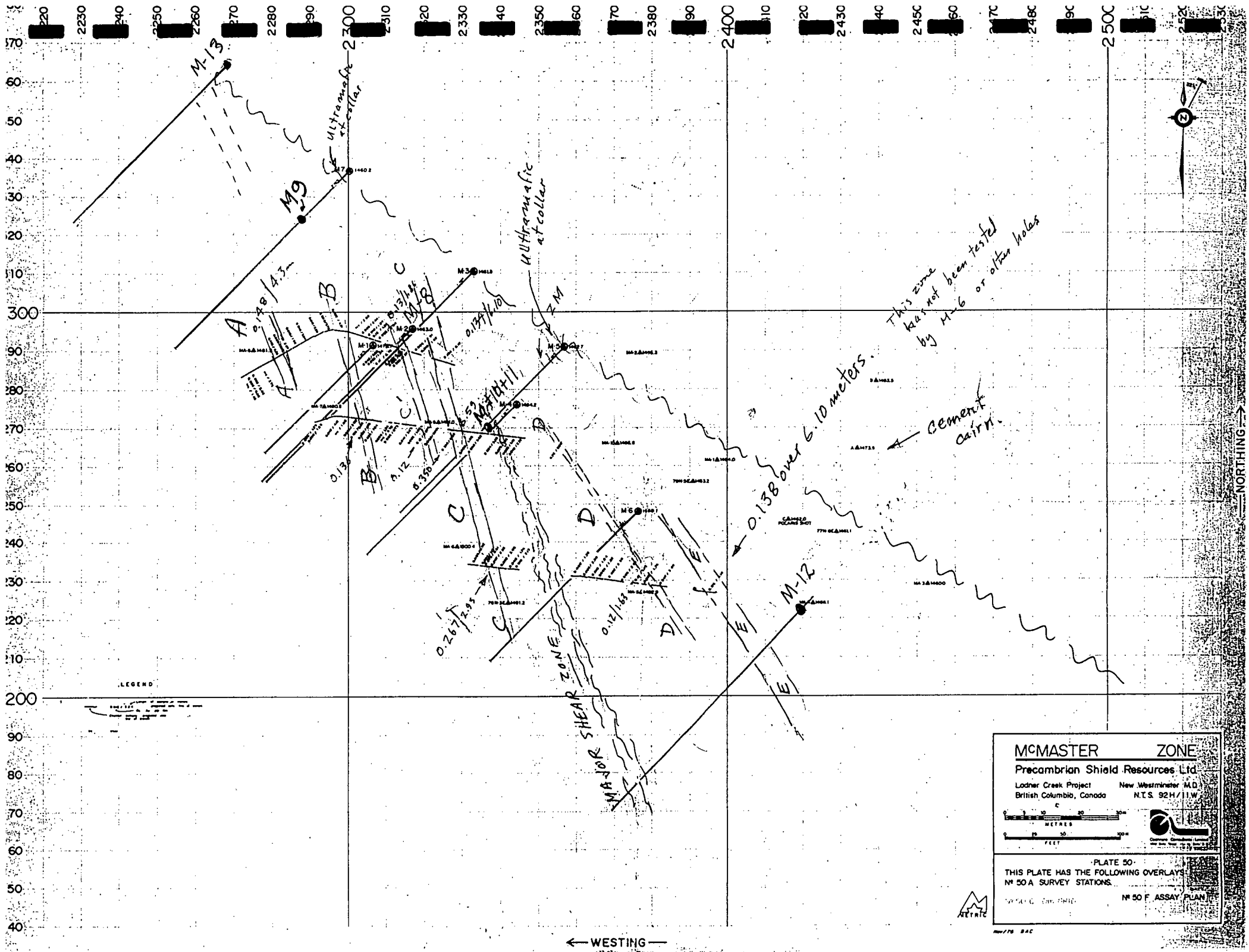
The mineralized zones at McMaster can be subdivided from west to east (refer to Figure 9) as follows:

Zone A: Exposed in North Trench - 0.148 oz/ton Au over 4.3 m. Intersected in hole M-7, M-9 and M-13. Strike length of about 100 metres. A Zone is open to the south.

- Zone B: Exposed in Middle Trench - 0.136 oz/ton Au over 2.53 m. Intersected in hole M-1, strike length of about 40 metres. B Zone is open to the south.
- Zone C: Highest Grade Zone, exposed in Middle and South trenches - 0.350 oz/ton over 6.59 m and 0.267 oz/ton over 2.93 m. Intersected in holes M-2, M-3, M-10, M-11 but is faulted off above holes M-4, M-5 and M-6. Strike length about 120 metres (as presently known)
- Zone D: Exposed in South Trench, 0.047 oz/ton Au over 3.60 metres. Intersected in hole M-6 and M-12. Strike length about 100 metres. D Zone is open to the south.
- Zone E: Exposed in South Trench (now sloughed in), 0.138 oz/ton Au over 6.10 metres. Intersected in hole M-12. 6.50 metres averaged 0.152 oz/ton gold. Strike length as presently known is 50 metres but is open to south.
- Note: Other mineralized zones can be expected to the west at depth such as indicated in holes M-9, M-12 and M-13.

The mineralized zones strike about 320° to 340° and dip 60° to the east. The trend of the bedding and mineralized zones appears to be dragged toward the McMaster Pond Fault due to right lateral movement.

The results of the mapping indicate that the McMaster Zone is underlain by the same stratigraphic package of rocks found in the Idaho Mine area (refer to Figure 8) as shown below.



M-13

M-9

M-3

M-2

M-1

M-4

M-5

M-6

M-7

M-8

M-10

M-11

M-12

ultramafic at collar

ultramafic at collar

A 0.28/4.3

B 0.13/1.4

C 0.13/1.4

D 0.13/1.4

E 0.13/1.4

F 0.13/1.4

G 0.13/1.4

H 0.13/1.4

I 0.13/1.4

J 0.13/1.4

K 0.13/1.4

L 0.13/1.4

M 0.13/1.4

N 0.13/1.4

O 0.13/1.4

P 0.13/1.4

Q 0.13/1.4

0.13

0.12

0.355

0.267/2.95

0.12/1.83

0.138 over 6.10 meters

MA 34400

MA 34401

MA 34402

MA 34403

MA 34404

MA 34405

MA 34406

MA 34407

MA 34408

MA 34409

MA 34410

MA 34411

MA 34412

This zone has not been tested by M-6 or other holes

Cement cairn

MAJOR SHEAR ZONE



map 176 BAC

**SCHEMATIC
STRATIGRAPHIC COLUMN**

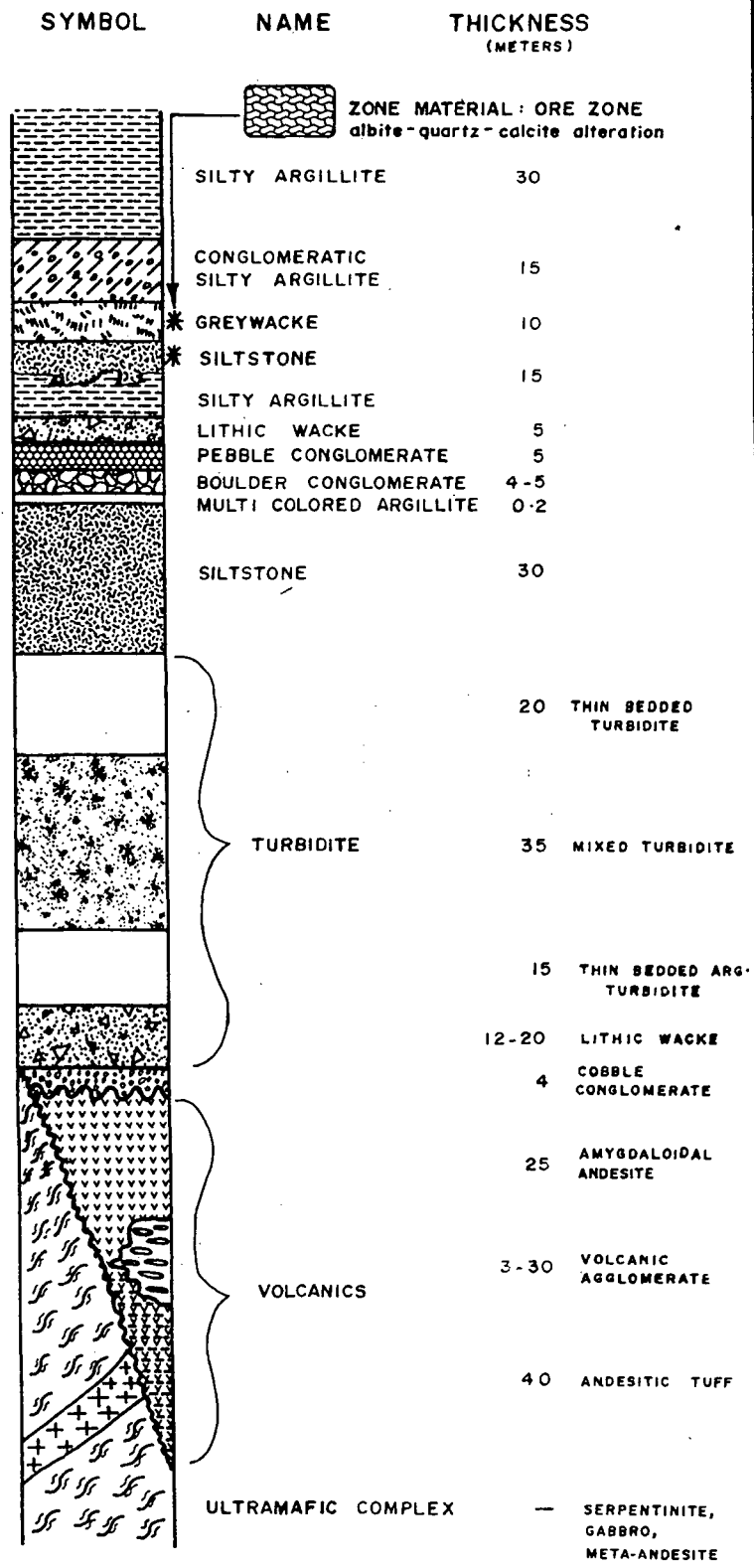


FIGURE 8 STRATIGRAPHY in the vicinity of the CAROLIN MINE.

TABLE II
STRATIGRAPHIC - STRUCTURAL PACKAGE
AT THE
McMASTER ZONE
(structurally "up" to the east)

- 1) Mylonitic Volcanics (Spider Peak Formation) (displaced along fault)
- 2) McMaster Pond Fault Structures and Splays (Serpentinite)
- 3) Variable: Chloritic, Greywacke, Lithicwacke
- 4) Zone Material - Mineralized Zone containing quartz-albite-carbonate alteration and arsenopyrite, pyrite, pyrrhotite and \pm chalcopyrite mineralization
- 5) Chloritic Greywacke
- 6) Siltstone
 - (a) Siltstone
 - (b) Argillaceous Siltstone
- 7) Turbidite
 - (a) Greywacke
 - (b) Lithicwacke)
 - (c) Conglomerates (pebble to boulder)) graded sequences
 - (d) Siltstone)
- 8) Conglomeratic Argillite (matrix supported clasts)
- 9) Mylonitic Volcanics
- 10) Hozameen Fault and Ultramafic Complex

The mineralized zones exhibit the same sulfide mineralization and alteration found in the "Zone Material" of the Idaho ore bodies. Pyrite, arsenopyrite and pyrrhotite predominate, but minor chalcopyrite was observed. The alteration assemblage is pervasive albite-quartz and carbonate.

The rocks at McMaster strike northwesterly, ranging between 300° and 318° on the west part of the trenches, and dip steeply to the northeast between 65° and 80°. Several large, westerly dipping shear zones having been identified trending subparallel to the mineralized zones. These shear zones appear to truncate the mineralized zones at depth which gives an overall package of thin fault wedges.

Individual rock types in the general McMaster area can be subdivided as follows:

McMaster Zone Rock Types

- 1) **Myolinitic Volcanics:** found on the east side of the McMaster Valley. These volcanic rocks are characterized by a fine grained greenish highly brecciated, chloritic andesite or basalt.
- 2) **McMaster Pond Fault Structure:** a northwesterly trending large fault structure that floors the McMaster Valley. This fault structure is filled with an elongate serpentinite body as indicated by the ground magnetometer results.
- 3) **Serpentinite:** this unit has been observed occasionally at the collars of drill holes located at the base of the McMaster ridge. The serpentinite is characterized by dark green to black highly sheared (slickensided), serpentinite cut by white calc-silicate stringers and occasional antigorite veinlets.
- 4) **Zone Material:** this rock type is an alteration feature consisting of albite, quartz and carbonate. The rock is characterized by a light grey (may vary to dark charcoal grey) colour that is often cut by coarser quartz-carbonate-albite veins. Pervasive quartz-carbonate-albite flooding is also common. The rock is often well fractured with calcite occurring on vein and fracture margins. Albite is also present as discrete white crystals occurring within the more translucent quartz veins. A dramatically increased level of sulfide mineral in the range of 10 - 20% by volume clearly distinguished "Zone Material" alteration from other altered units. Pyrite, pyrrhotite, arsenopyrite and occasionally chalcopyrite are the primary sulfides. Gold is found associated with this mineral assemblage. The sulfides occur as blebs within veins, coatings along fractures and vein margins and disseminations.
- 5) **Chloritic Greywacke:** dark green grey, fine grained, massive. Generally found adjacent to Zone Material sections. Chlorite alteration is intense and chloritic laminations are found on slickensided surfaces in sheared areas within this unit. Quartz alteration is normally weak. Carbonate alteration can vary considerably and occurs mainly along fine hairline fractures and as fine stringers.

- 6) **Siltstone:** in the McMaster Zone two distinct siltstone units were found.
 - (a) **Siltstone:** grey to greenish grey, very fine grained well bedded unit. There is not change in grain size within the various layers. This sequence can take on a massive appearance when fractured.
 - (b) **Argillaceous Siltstone:** dark charcoal grey, very fine grained well layered unit. Slight variation in colour of individual layers in parts a strong banded appearance to rock. Graphite is usually abundant along bedding planes and on slickensided surface. Graphite appears to develop readily even in weakly sheared argillaceous siltstones. Carbonate alteration varies considerably and occurs pervasively throughout rock when alteration is strong.

- 7) **Turbidite:** this sequence of rocks consists of several distinct rock types that exhibit gradational contacts to each other commonly within one bed. The constituent sub units are:
 - (a) **Greywacke:** a finely clastic greenish grey unit that ranges from relatively even grained to well bedded appearance. The beds differ from siltstone in that grain size gradations are readily observed in layers less than 1 cm thick. Colour ranges from light grey green to dark greenish grey.
 - (b) **Lithicwacke:** light grey green unit composed of angular elongated clasts. The coarse grained lithicwacke sequence grade in to pebble conglomerates. The lithicwacke units generally have gradational contacts between fine to medium to coarse grained sections although abrupt contacts are observed. Framework grains are always close packed. Alignment of clasts imparts a rough pseudofoliated appearance.
 - (c) **Pebble Conglomerates:** Pebble conglomerates are generally characterized by a light grey colour and clast size of less than 2 cm diameter. Clasts are close packed and are often flattened. They form the basalt part of the lithicwacke units.

(d) **Siltstone:** light green grey, thin bedded to laminated sections. Gradational grain sizing within layers is not common.

- 8) **Conglomeratic Argillite:** this unit occurs very commonly in the McMaster Zone area and is a dark charcoal grey coloured sequence. A distinguishing feature is that it is most commonly found as a very loosely packed unit with pebble sized clasts. Dark grey matrix material surrounds most clasts. This unit occurs to the east of the Idaho No. 1 ore zone at 934N.

DIAMOND DRILLING - 1975 AND 1989

In November 1975, a total of 1,699 feet of surface diamond drilling was completed in seven holes. This core was relogged in September 1989, enabling accurate correlation between the recent surface mapping and the subsurface data base. It is a credit to Carolin Mines Ltd. that the 1975 McMaster core was available and in good shape.

In October - November, a further 1,369 feet of diamond drilling was completed in six holes. Drilling has been concentrated on six drill sections: Figures 10 - along hole M-13, Figure 11 - along holes M-7 and 9, Figure 12 - along holes M-1, 2, 3 and 8, Figure 13 - along holes M-4, 5, 10 and 11, Figure 14 - along holes M-6, Figure 15 - along M-12.

Significant mineralized zones are listed in Table III.

On the northern most cross-section (Figure 10), four separate mineralized zones were encountered. Hole M-13 was collared in "A" Zone which is exposed in the new road cut. The other three mineralized zones have not been noted in outcrop, but would be expected to subcrop to the west of the presently trenched area. Future diamond drilling should be done both above and below M-13.

Drill hole M-9 was placed 15 metres above hole M-7 (Figure 11). The zone encountered in M-9 is considerably higher grade than the zone in M-7. Faulting appears to have disrupted the section in the M-7 area as indicated by the density of slickensides and rubbly-broken core.

Four holes have been drilled under the Central Trench (Figure 12). Hole M-1 was collared west of the higher grade Zone "C", and only intersected a narrow part of Zone B. Holes M-2, M-8 and M-3 cut a faulted section of Zone "C" indicating a down-dip extent of 40 metres. Hole M-8 appears to have travelled mainly along a subsidiary west-dipping shear splay related to McMaster Pond Fault. The deeper parts of the holes are less disrupted by faulting and the conglomeratic argillite and turbidite units can be traced throughout the section between holes.

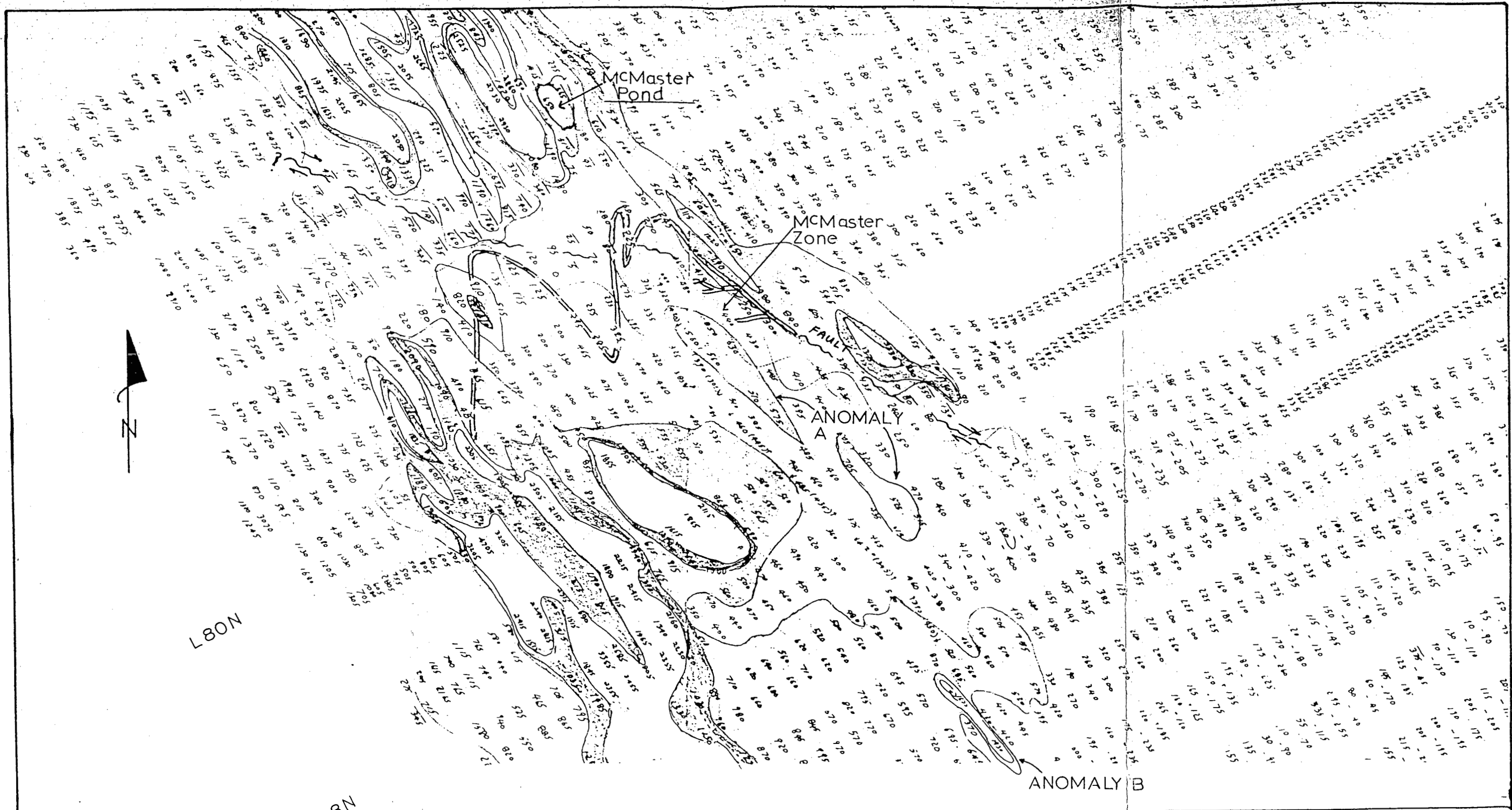
Several small fault slices are evident near the top of holes M-10 and 11 (Figure 13). Short intervals of conglomeratic argillite are associated with argillaceous siltstone, chloritic greywacke and weak zone material farther down the holes. This is in contrast to the turbidite assemblage encountered near the end of M-4 and 5. The wide brecciated fault structure noted in holes M-4 and 5 correlates well with the shear zone mapped in the south and central trenches. Part of Zone "C" was cut in holes M-10 and 11. A narrow mineralized zone was found in M-4 and 5 (Zone "D") which reflects the northern continuation of the Zone "D" exposed in the South Trench.

Hole M-6 intersected a narrow part of Zone "D" (Figure 14). The length of the surface drill rig did not allow a hole to be collared in the South Trench to test Zone "C". Future drilling will require a small drill rig capable of drilling 30 to 50 metre holes.

The most southerly hole (M-12, Figure 15) intersected 6.5 metres averaging 0.152 oz/ton Au. This is Zone "E" which was cut in the South Trench (0.138 oz/ton Au over 6.10 m) but is now covered by slough from the top of the trench. This zone is the most immediate major target for future drilling. To the south, Zone "E" would be expected to diverge from the McMaster Pond Fault structures, thus increasing the possibility of down-dip continuity. Accurate mapping of the cross fault, sub-parallel but 120 m south of the McMaster Pond Fault, will determine the details of the follow-up program to trace Zone "E" toward the south.

TABLE III
SIGNIFICANT GOLD INTERSECTIONS AT McMASTER ZONE
1975 and 1989 DIAMOND DRILLING

Drill Hole	Drill Intersections (m)	Drill Intersections (ft)	Length		Average Grade oz/ton Au
1975					
M-1	12.65 - 16.80 m	41.50 - 55.12 ft.	4.15 m	(13.6 ft)	0.070
M-2	5.80 - 11.53 m	19.03 - 37.83 ft.	5.93 m	(19.45 ft)	0.130
M-3	19.90 - 26.26 m	65.29 - 86.15 ft.	6.36 m	(20.86 ft)	0.110
	62.40 - 64.0 m	204.72 - 209.97 ft.	1.90 m	(6.23 ft)	0.070
M-4	6.32 - 9.70 m	20.73 - 31.82 ft.	3.38 m	(11.09 ft)	0.069
M-5	24.75 - 26.14 m	81.20 - 85.76 ft.	1.39 m	(4.56 ft)	0.050
M-6	6.70 - 12.70 m	21.98 - 41.67 ft.	6.00 m	(19.68 ft)	0.064
M-7	26.71 - 40.00 m	87.63 - 131.23 ft.	13.29 m	(43.60 ft)	0.045
1989					
M-8	10.19 - 15.60 m	33.43 - 51.13 ft.	5.41 m	(17.75 ft)	0.047
M-9	2.74 - 8.48 m	8.99 - 27.82 ft.	5.74 m	(18.83 ft)	0.083
	8.48 - 12.00 m	27.82 - 39.37 ft.	3.52 m	(11.55 ft)	0.035
	12.00 - 22.61 m	39.37 - 74.18 ft.	10.61 m	(34.81 ft)	0.076
	27.4 - 22.61 m	8.99 - 74.18 ft.	19.87 m	(65.20 ft)	0.070
	83.41 - 84.41 m	273.65 - 276.93 ft.	1.00 m	(3.29 ft)	0.090
M-10	2.28 - 6.05 m	7.48 - 19.85 ft.	3.77 m	(12.37 ft)	0.132
M-11	2.44 - 8.74 m	8.00 - 28.67 ft.	6.29 m	(20.63 ft)	0.0677
	8.74 - 17.00 m	28.67 - 55.77 ft.	8.27 m	(27.10 ft)	0.015
	44.69 - 48.09	146.62 - 157.77 ft.	3.4 m	(11.15 ft)	0.023
M-12 includes	8.50 - 15.00 m	27.89 - 49.21 ft.	6.5 m	(26.25 ft)	0.152
	12.50 - 15.00 m	41.01 - 49.21 ft.	2.5 m	(8.20 ft)	0.190
	51.5 - 54 m	168.96 - 177.16 ft.	2.5 m	(8.20 ft)	0.053
M-13	0.91 - 3.00 m	2.98 - 9.84 ft.	2.09	(6.85 ft)	0.124
	3.00 - 12.50 m	9.84 - 41.01 ft.	9.5 m	(31.17 ft)	0.037
	12.50 - 14.34 m	41.01 - 47.04 ft.	1.84 m	(6.03 ft)	0.160
	0.91 - 14.34 m	2.90 - 47.04 ft.	13.43 m	(44.06 ft)	0.067
	25.60 - 28.00 m		2.4 m	(7.87 ft)	0.043
	30.28 - 40.00 m		9.72 m	(31.89 ft)	0.032



660 - RELATIVE MAG. VALUES (Gammas)

FAULT ZONE
CONTOUR INTERVAL : 500 gammas

CAROLIN MINES LTD.	
MCMaster PROJECT	
MAGNETOMETER SURVEY-1975	
SCALE: 1" = 500'	FIGURE 17
1: 6000	

GEOCHEMISTRY

Soil sampling in 1975 outlined very highly anomalous values (in excess of 1,450 parts per billion (ppb), gold) over an area 500 feet long by 100 to 200 feet in width (Figure 16). In addition, anomalous soil results give a well defined (greater than 540 ppb Au) pattern over 1,700 feet in length from line 72N+500E to Line 86N+00E. Lower value gold-in-soil results (greater than 90 ppb Au) continue southeast toward the Montana Adit above the Idaho #2 zone outcrop. The size and intensity of the soil anomaly over the McMaster Zone is similar to the soil anomaly found over the Idaho Zone.

In 1986, follow-up soil sampling was completed by Arctex Engineering Services for Pennant Holdings Ltd. over eleven small grids between the Aurum Zone and the Pipestem Mine. The 72+00N, 5+00E grid is located southeast of the McMaster Zone. Gold values in soil ranged from 10 ppb to 1250 ppb. This grid should be extended to the east to the Rush of the Bull showings in conjunction with excavator.

GEOPHYSICS

Ground Magnetometer Survey (1975)

A wide-spaced ground magnetometer survey covering the McMaster area was conducted during 1975 as part of a much larger program. The magnetometer results indicate the northwesterly trend of the Coquihalla Serpentine Belt. This ultramafic body is bounded by the East Hozameen Fault. The magnetic pattern shows a major dislocation in the strike continuity of the Belt in the McMaster Zone area. The northwest trending general McMaster Pond Fault that occupies the McMaster - Upper Deadman Creek Valley displaces the serpentine 800 metres to the east by right-lateral strike-slip motion.

On a smaller scale, northeast to southwest cross faulting has moved a segment of the serpentinite body eastwards between Line 69N and Line 80N. This cross faulting may have an impact on the continuity of the McMaster Zone mineralization to the southeast.

In the McMaster Zone mineralized area along the ridge west of the cut road (Figure 17) two discrete lower intensity anomalies designated Anomaly A were outlined. These anomalies are probably caused, in part, by pyrrhotite content of argillaceous siltstones, turbidites and conglomeratic argillites adjacent to the gold-bearing quartz - albite - carbonate zones.

A narrow high intensity anomaly (Anomaly B) is located approximately 600 metres southeast of the end of the trench access road. This anomaly may represent a local shear zone which could be associated with mineralized zone material.

CONCLUSIONS

The 1989 work program on the McMaster Zone was successful in correlating the newly collected surface mapping data with the limited subsurface data from the 1975 diamond drilling to form a coherent geological synthesis. This new geological interpretation was tested and extended by a short six hole diamond drill program in October, 1989. The continuity of certain mineralized zones between sections (and to depth) within particular fault wedges has been established.

The five known outcropping zones at McMaster strike about 320° to 340° and dip 60° E. They are truncated at depth by a series of 75° west-dipping post-mineralization faults. Zone C appears to extend to a depth of about 40 metres along cross section M-1, 2, 3 and 8. The Southern (Zone E) and Northern (Zone A) limits of the McMaster Zone are open and the high gold-in-soil results suggest a possible extension 500 feet to the northwest and at least 600 feet to the southeast. The magnitude and scope of the future work required to fully evaluate the McMaster Zone can now be accurately estimated.

Since the McMaster Zone outcrops along a small knoll, the extraction by open cast methods of near surface ore (that may be defined by future programs) is a distinct possibility.

Considering the possible strike length, number of mineralized zones, width of mineralization, gold grades and known extent down-dip and general geological parameters, in my opinion, the McMaster Zone has the potential to contain a mineral deposit similar in size and grade to the Idaho Zone which was developed into the Carolin Mine. (Approximate published reserves at the start of mining of 1.5 million tons averaging 0.141 oz/ton at a 0.08 cut-off with 20% dilution.)

RECOMMENDATIONS

Based on the preliminary results of the 1989 work program, the following orderly exploration program is recommended:

1. Accurately cut lines and conduct 1:1000 geological mapping fill-in-soil sampling and ground magnetometer surveys of the area between lines 68N to L89N (50 metre line spacing with 20 metre stations). Special attention should be given to the old trenches on Line L73N near (west) the "Rush of the Bull" Showing.
2. Road (Tote road) construction by a tracked excavator (Cat 225 or JD 790) south from 77N (present road) to L68N, a distance of 1,000 feet and north to L86N. An excavator operator experienced in mountainous terrain and pioneering roads is essential for this phase of the program. Consideration of paying a premium for the right operator should be entertained.
3. Trenching by tracked excavator at 100 foot intervals following the McMaster Mineralized Zones to the south. Careful mapping and channel / or chip sampling of mineralization encountered is required.
4. 6,000 feet of diamond drilling split between a small (Gopher-type drill capable of 200-300' holes) drill and larger drill capable of 500' to 1,000' holes.

The cost of such a program will be approximately \$353,000 (Canadian). (Refer to Cost Estimate on page 19.) If this program is successful in extending the continuity of the McMaster Zones to the north and south and to depth, then a detailed major definition drill program would be required to define mineable ore reserves.

Respectfully submitted,

J.T. Shearer, M.Sc., FGAC
New Global Resources Ltd.

COST ESTIMATE
FOR FUTURE WORK
McMASTER ZONE

PHASE 1990-1, JUNE 1 TO AUGUST 15, 1990

1)	Geological mapping, 1:1000, Grid Control	\$ 16,000
2)	Transit-EDM Survey Control	8,000
3)	Road Building (Tracked Excavator), 800 m	22,000
4)	Trenching (Tracked Excavator) at 30 m intervals	28,000
5)	Mob & Demob of Excavator	2,000
6)	Camp Costs (food & supplies) utilizing McMaster Camp	4,000
7)	Transportation	3,000
8)	Analytical (rock and soil)	8,000
9)	Compilation and Report Preparation	<u>4,000</u>
	Sub-total	95,000
10)	Contingencies 10%	<u>9,500</u>
	Sub-total	104,500

PHASE 1990-2, AUGUST 1 TO SEPTEMBER 30, 1990

1)	Diamond Drilling, 6,000 ft. at \$35/ft (all in) plus geological supervision	210,000
2)	Analytical	10,000
3)	Compilation and Report Preparation	<u>6,000</u>
	Sub-total	226,000
	Contingencies 10%	<u>23,000</u>
		<u>249,000</u>

GRAND TOTAL \$ 353,500

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APPENDIX I

STATEMENT OF QUALIFICATIONS

STATEMENT OF QUALIFICATIONS

I, Johan T. Shearer of the City of Port Coquitlam, in the Province of British Columbia, do hereby certify:

1. I graduated in Honours Geology (B. Sc. 1973) from the University of British Columbia and the University of London, Imperial College, (M. Sc. 1977).
2. I have practised my profession as an Exploration Geologist continuously since graduation and have been employed by such mining companies as McIntyre Mines Ltd., J.C. Stephen Explorations Ltd., Carolin Mines Ltd. and TRM Engineering Ltd. I am presently employed by New Global Resources Ltd.
3. I am a fellow of the Geological Association of Canada (Fellow No. F439). I am also a member of the Canadian Institute of Mining and Metallurgy, the Geological Society of London and the Mineralogical Association of Canada.
4. I supervised all exploration on the Ladner Creek North Project from February 1981 to November 1982 and worked underground as exploration geologist at the Idaho Mine from November 1982 to February 1984 engaged in detail geological mapping and project supervision. I have logged diamond drill core and supervised the geological staff during the mapping and relogging and drilling in August to November 1989 for the McMaster Program.
5. I have no interest in Carolin Mines Ltd. or any of its affiliated companies, nor do I expect to receive any in the future. I consent to the use of this report in or in connection with a prospectus or in a statement of Material facts relating to the raising of funds.
6. Other New Global personnel working on the McMaster Project were B. Lennan and W. Howell, both graduate geologists. A further note on these individuals is attached.

Dated at Vancouver, British Columbia

J.T. Shearer, M. Sc., F.G.A.C.
December 6, 1989

APPENDIX II

STATEMENT OF COSTS (McMASTER 1989 PROGRAM)

STATEMENT OF COSTS
McMASTER ZONE
1989 EXPLORATION PROGRAM

Labour, Wages and Benefits

J.T. Shearer, M.Sc., Senior Geologist 22.5 days at \$300 per day	\$ 6,750.00
W.B. Lennan, B.Sc., Project Geologist 39 days at \$250 per day	9,750.00
W.A. Howell, B.Sc., Geologist 6 days at \$250 per day	1,500.00
S.C. Shearer, Line Cutter, Sampler and Core Splitter 17 days at \$140 per day	<u>2,380.00</u>
Sub-total	20,380.00

Expenses

Truck rental	
Redhawk	1,170.00
New Global	778.80
Meals and groceries	1,419.51
Accommodation - motel	504.94
Gasoline	755.53
Camp supplies and survey supplies	111.33
Consultant expenses (W.A. Howell) includes truck rental, gas and food	260.94
Drafting supplies (mylar)	43.25
Reproduction	80.23
Telephone (estimated portion to McMaster)	100.00
Assaying (Chemex Labs, 295 samples at \$13.50/sampl)	,3982.50
Diamond drilling (369 ft. x \$22.40/ft.)	30,665.60
Cat work on road (estimated portion to McMaster)	<u>1,200.00</u>
Sub-total	41,073.23

Total

\$61,453.23

APPENDIX III

DIAMOND DRILL CONTRACT

APPENDIX IV

LIST OF PERSONNEL AND DATES WORKED

LIST OF PERSONNEL AND DATES WORKED
ON 1989 McMASTER ZONE

<u>Name</u>	<u>Occupation</u>	<u>Address</u>	<u>Dates Worked</u>
J.T. Shearer	Senior Geologist (M.Sc.)	3832 St. Thomas St. Port Coquitlam, B.C.	Aug 9; Sept 1, 27, 28, 29, 30; Oct 2, 10, 17, 18, 20, 24, 26, 27, 30, 31; Nov 1, 5, 7, 8, 9, 10(½ day), 13, 15 Total 22.5 days
W.B. Lennan	Project Geologist (B.Sc. 1973)	876 Lynwood Ave. Port Coquitlam, B.C.	Sept 8, 23, 24(½ day), 25, 29; Oct 10-20, 23-31; Nov 1-9, 10(½ day), 13-15 Total 39 days
W.A. Howell	Geologist (B.Sc.)	15294 - 96A Ave. Surrey, B.C.	Oct 25-30 Total 6 days
S.L. Shearer	Line Cutter Core Splitter	3345 Mason Ave. Port Coquitlam, B.C.	Sept 25-29; Oct 30, 31; Nov 1-9, 15 Total 17 days

Mr. Lennan graduated from the University of British Columbia in 1973. He has worked continuous as a senior geologist for a number of major companies since graduation.

Mr. Howell graduated from the University of British Columbia in 1972. He has worked continuous in mineral exploration since that time.

APPENDIX V

ANALYTICAL PROCEDURE AND ASSAY CERTIFICATES



Chemex Labs Ltd.

Analytical Chemists • Geochemists • Registered Assayers

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

PHONE (604) 984-0221

T. ARLO... LINE... MIT...

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project : MCMASTER
Comments : CC: J. SHEARER

*Page :
Tot. Pages: 3
Date : 27-NOV-89
Invoice # : I-8930533
P.O. # :

CERTIFICATE OF ANALYSIS A8930533

SAMPLE DESCRIPTION	PREP CODE	Au oz/T									
73051	207	---	0.020								
73052	207	---	0.014								
73053	207	---	0.048								
73054	207	---	0.129								
73055	207	---	0.004								
73056	207	---	0.008								
73057	207	---	0.002								
73058	207	---	0.002								
73059	207	---	0.032								
73060	207	---	0.008								
73061	207	---	0.010								
73062	207	---	0.010								
73063	207	---	0.004								
73064	207	---	< 0.002								
73065	207	---	0.002								
73066	207	---	0.002								
73067	207	---	0.002								
73069	207	---	0.002								
73070	207	---	< 0.002								
73071	207	---	0.002								
73072	207	---	< 0.002								
73073	207	---	0.002								
73074	207	---	>> 0.002								
73075	207	---	>> 0.002								
73076	207	---	0.006								
73077	207	---	< 0.002								
73078	207	---	0.139								
73079	207	---	0.185								
73080	207	---	0.066								
73081	207	---	0.104								
73082	207	---	0.062								
73083	207	---	0.060								
73084	207	---	0.008								
73085	207	---	0.008								
73086	207	---	0.044								
73087	207	---	0.032								
73088	207	---	0.026								
73089	207	---	0.014								
73090	207	---	0.018								
73091	207	---	0.026								

CERTIFICATION : J. Gussel



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PHONE (604) 984-0221

TO: ROL NES SITE

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project : MCMASTER
Comments : CC: J. SHEARER

Page : 2
Tot. Pages: 3
Date : 27-NOV-89
Invoice # : I-8930533
P.O. # :

CERTIFICATE OF ANALYSIS A8930533

SAMPLE DESCRIPTION	PREP CODE	Au oz/T									
73092	207 ---	0.046									
73093	207 ---	0.068									
73094	207 ---	0.030									
73095	207 ---	0.010									
73096	207 ---	0.024									
73097	207 ---	0.062									
73098	207 ---	0.082									
73099	207 ---	0.026									
73100	207 ---	0.056									
73101	207 ---	0.167									
73102	207 ---	0.208									
73103	207 ---	0.158									
73104	207 ---	0.082									
73105	207 ---	0.018									
73106	207 ---	0.004									
73107	207 ---	0.006									
73108	207 ---	0.008									
73109	207 ---	0.016									
73110	207 ---	0.006									
73111	207 ---	0.038									
73112	207 ---	0.018									
73113	207 ---	0.044									
73114	207 ---	0.008									
73115	207 ---	0.024									
73116	207 ---	0.004									
73117	207 ---	0.016									
73118	207 ---	0.038									
73119	207 ---	0.046									
73120	207 ---	0.038									
73121	207 ---	0.010									
73122	207 ---	0.072									
73123	207 ---	0.008									
73124	207 ---	0.006									
73125	207 ---	0.034									
73126	207 ---	0.052									
73127	207 ---	0.004									
73128	207 ---	0.042									
73129	207 ---	0.028									
73130	207 ---	0.036									
73131	207 ---	0.034									

CERTIFICATION : *Alhnsi*



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 PHONE (604) 984-0221

THE NEW GOLD L. RESOURCES

548 BEATTY ST.
 VANCOUVER, BC
 V6B 2L3

Project : CAROLIN (MCMASTER)
 Comments :

Page # :
 Tot. Pages : 1
 Date : 5-NOV-89
 Invoice # : I-8929436
 P.O. # : NONE

CERTIFICATE OF ANALYSIS A8929436

SAMPLE DESCRIPTION	PREP CODE	Au oz/T RUSH							
73401	236 ---	0.078							
73402	236 ---	0.040							
73403	236 ---	0.070							
73404	236 ---	0.124							
73405	236 ---	0.050							
73406	236 ---	0.193							
73407	236 ---	0.020							
73409	236 ---	0.030							

CERTIFICATION : *W. St. Amant*



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To: CAROLIN MINES LIMITED

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project: MMMASTER
Comments: CC: J T SHEAKER

**Page No.: 1
Tot. Pages: 1
Date: 3-NOV-89
Invoice #: I-8929415
P.O. #: NONE

CERTIFICATE OF ANALYSIS A8929415

SAMPLE DESCRIPTION	PREP CODE	Au g./T RUSH FA							
73408	236	---	0.046						
73410	236	---	0.060						
73411	236	---	0.032						
73412	236	---	0.030						
73413	236	---	0.126						
73414	236	---	0.159						
73415	236	---	0.066						
73416	236	---	0.086						
73417	236	---	0.044						
73418	236	---	0.071						
73419	236	---	0.070						
73420	236	---	0.050						
73421	236	---	0.052						
73422	236	---	0.052						
73423	236	---	0.046						
73424	236	---	0.008						
73425	236	---	0.006						
73426	236	---	0.004						
73427	236	---	0.012						
73428	236	---	0.006						
73429	236	---	0.012						
73430	236	---	0.026						
73431	236	---	0.003						
73432	236	---	0.003						
73433	236	---	0.003						
73434	236	---	0.003						
73435	236	---	0.003						
73436	236	---	0.003						
73437	236	---	0.008						
73438	236	---	0.003						

CERTIFICATION

Alvin



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PHONE: (604) 984-0221

T. CAROLAN MINES LIMITED

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project: MCMASTER

Comments: CC: J. T. SHEARER

Page No. 1
Tot. Pages: 2
Date: 20-NOV-89
Invoice #: I-8930335
P.O. #

CERTIFICATE OF ANALYSIS A8930335

SAMPLE DESCRIPTION	PREP CODE		Au FA									
			oz/T									
73439	214	---	0.012									
73440	214	---	0.020									
73441	214	---	0.008									
73442	214	---	< 0.003									
73443	214	---	< 0.003									
73444	214	---	0.018									
73445	214	---	0.010									
73446	214	---	0.012									
73447	214	---	0.022									
73448	214	---	0.016									
73449	214	---	0.014									
73450	214	---	0.014									
73451	214	---	0.014									
73452	214	---	0.010									
73453	214	---	0.018									
73454	214	---	0.090									
73455	214	---	0.014									
73456	214	---	0.008									
73457	214	---	< 0.003									
73458	214	---	0.008									
73459	214	---	< 0.003									
73460	214	---	0.004									
73461	214	---	0.174									
73462	214	---	0.102									
73463	214	---	0.034									
73464	214	---	0.160									
73465	214	---	0.016									
73466	214	---	0.163									
73467	214	---	0.032									
73468	214	---	0.006									
73469	214	---	0.004									
73470	214	---	0.008									
73471	214	---	0.014									
73472	214	---	0.018									
73473	214	---	0.006									
73474	214	---	< 0.003									
73475	214	---	0.004									
73476	214	---	< 0.003									
73477	214	---	0.013									
73478	214	---	< 0.003									

CERTIFICATION : W. Sturmon



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TERRACOL MINES LIMITED

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VANCOUVER, BC
V6C 1G8

Project : MCMASTER

Comments: CC: J.T. SHEARER

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Tot. Pages: 2
Date : 20-NOV-89
Invoice # : I-8930335
P.O. #

CERTIFICATE OF ANALYSIS A8930335

SAMPLE DESCRIPTION	PREP CODE	Au FA oz/T								
73479	214 ---	< 0.003								
73480	214 ---	0.008								
73481	214 ---	0.004								

CERTIFICATION : W. Santomasini



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PHONE (604) 984-0221

To: CAROLIN MINES LIMITED

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project: MCMASTER

Comments: CC: NEW GLOBAL RESOURCES

Page No. : 1
Tot. Pages: 2
Date : 14-NOV-89
Invoice # : I-8929830
P.O. # : NONE

CERTIFICATE OF ANALYSIS A8929830

SAMPLE DESCRIPTION	PREP CODE		Au FA										
			oz/T										
73482	207	---	0.012										
73483	207	---	0.026										
73484	207	---	0.024										
73485	207	---	< 0.003										
73486	207	---	< 0.003										
73487	207	---	< 0.003										
73488	207	---	0.020										
73489	207	---	0.006										
73490	207	---	< 0.003										
73491	207	---	< 0.003										
73492	207	---	0.006										
73493	207	---	0.012										
73494	207	---	0.008										
73495	207	---	< 0.003										
73496	207	---	< 0.003										
73497	207	---	0.004										
73498	207	---	0.012										
73499	207	---	<< 0.003										
73500	207	---	<< 0.003										
73519	207	---	<< 0.003										
73520	207	---	0.011										
73521	207	---	< 0.003										
73522	207	---	0.006										
73523	207	---	< 0.003										
73524	207	---	0.008										
73525	207	---	<< 0.003										
73526	207	---	0.003										
73527	207	---	0.010										
73528	207	---	0.018										
73529	207	---	0.072										
73530	207	---	0.081										
73531	207	---	0.050										
73532	207	---	0.032										
73533	207	---	0.020										
73534	207	---	0.004										
73535	207	---	< 0.003										
73536	207	---	0.139										
73537	207	---	0.104										
73538	207	---	0.086										
73539	207	---	0.028										

CERTIFICATION :

N. S. Morrison



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602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project: MCMASTER

Comments: CC: NEW GLOBAL RESOURCES

**Page No. : 2

Tot. Pages: 2

Date : 14-NOV-89

Invoice # : I-8929830

P.O. # : NONE

CERTIFICATE OF ANALYSIS A8929830

SAMPLE DESCRIPTION	PREP CODE		Au FA oz/T									
73540	207	---	0.038									
73541	207	---	0.056									
73542	207	---	0.012									
73543	207	---	0.016									
73544	207	---	0.008									
73545	207	---	0.028									
73546	207	---	0.124									
73547	207	---	0.193									
73548	207	---	0.012									
73549	207	---	0.018									
73550	207	---	0.016									
73551	207	---	0.028									
73552	207	---	0.016									
73553	207	---	0.008									
73554	207	---	0.012									
73555	207	---	0.010									
73556	207	---	< 0.003									
73557	207	---	0.006									
73558	207	---	< 0.003									
73559	207	---	0.004									
73560	207	---	0.006									
73561	207	---	< 0.003									
73562	207	---	0.010									
73563	207	---	<< 0.003									
73564	207	---	<< 0.003									
73565	207	---	0.014									
73566	207	---	0.004									
73567	207	---	0.006									
73568	207	---	0.012									
73569	207	---	0.022									
73570	207	---	0.010									
73571	207	---	0.010									
73572	207	---	0.008									
73573	207	---	0.004									
73574	207	---	0.006									
73575	207	---	< 0.003									
73576	207	---	0.008									
73577	207	---	0.010									

CERTIFICATION : *W. Spence*



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To CAROLIN MINES LIMITED

602 - 700 W. PENDER ST.
VANCOUVER, BC
V6C 1G8

Project : MCMASTER

Comments : CC: J.T. SHEARER

**Page No : 1

Tot. Pgs: 1

Date : 06-NOV-89

Invoice # : I-8929621

P.O. # : NONE

CERTIFICATE OF ANALYSIS A8929621

SAMPLE DESCRIPTION	PREP CODE	Au oz/T RUSH FA							
73501	236 ---	0.003							
73502	236 ---	0.008							
73503	236 ---	0.100							
73504	236 ---	0.056							
73505	236 ---	0.024							
73506	236 ---	0.164							
73507	236 ---	0.188							
73508	236 ---	0.198							
73509	236 ---	0.092							
73510	236 ---	0.108							
73511	236 ---	0.051							
73512	236 ---	0.030							
73513	236 ---	0.170							
73514	236 ---	0.264							
73515	236 ---	0.146							
73516	236 ---	0.182							
73517	236 ---	0.188							
73518	236 ---	0.012							

CERTIFICATION :

H. San Antonio

APPENDIX VI

DIAMOND DRILL LOGS: 1989 AND 1975 (RELOGGED)



LOCATION (LEVEL): M ^c MASTER ZONE (SURFACE)		PROJECT: M ^c MASTER		HOLE NUMBER: 89-M-8	
DIP: -55° On Section M-1, M-2 & M-3		DIAMOND DRILL RECORD		A2. 225°	
LATITUDE: N		LENGTH: 45.41m (149')		CLAIM NUMBER:	
DEPARTURE: E		CORE SIZE:		DATE LOGGED: Oct 28/89	
STARTED: Oct 25/89 at Noon		FINISHED: Oct 26/89 1AM		LOGGED BY: JTS, WBL.	
O.B. THICKNESS: 4.57m		STARTED: Oct 25, 1989 05		FINISHED: Oct 26, 1989 05	
B.R. THICKNESS: 45.41m		STARTED: Oct 25, 1989		FINISHED: Oct 26, 1989 05	
CONTRACTOR: Boiveneau		CORE STORED: IDAHO CORE SHACK		TOTAL RECOVERY: %	
				SURVEY: DEPTH: (496) 45.41m BEARING: 225° ANGLE: Reading 59.5° Correct 51°	

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: To TEST M-1, M-2, M-3 SECTION BETWEEN COMMENT: HOLES M-2 & M-3 * Driller estimated bedrock at 3.96 m. Ran casing to 4.57 m INTERVAL Bedrock (well defined) at 3.53 m from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	CALCITE	ALBITE						From	to		
0-3.96								0-3.96 Muddy Overburden							
3.96-7.01								3.96-7.01 CORED BOULDERS?							
7.01-8.53								7.01-8.53 CHLORITIC GREYWACKE - possibly boulders 8.53-8.55 very little core - weathered and ground up?							
8.53-10.19								8.53-10.19 ZONE MATERIAL - Qtz - carbonate - albite flooded Coarse lithic wacke to conglomeratic unit. At 9.4m drusy Qtz in open fracture 5-10% sulfide, mostly py. 1-2% sp. py.		73596	8.95	9.5	0.55	0.004	
10.19-10.86								10.19-10.86 ALTERED COARSE LITHIC WACKE - Intensely silicified, 1-3%		73597	9.50	10.19	0.69	0.006	
10.86-12.12								10.86-12.12 ALTERED ARGILLACEOUS SILTSTONE - Dark charcoal grey with Qtz - carbonate veins. Disseminated pyrrhotite.		73598	10.19	10.86	0.67	0.030	
12.12-14.0								12.12-14.0 ALTERED CONGLOMERATIC ARGILLITE - dark grey loosely packed primarily pebble conglomerate, weak to moderate silification 21% py		73599	10.86	11.86	1.00	0.042	
14.0-14.63								14.0-14.63 ALTERED GREYWACKE - chl. & graphite on foliation planes, 1% py		* 73051	11.86	13.00	1.14	0.020	
14.63-15.60								14.63-15.60 ALTERED COARSE LITHIC - strongly foliated 1-3% py Fault zone at 15.3 to 15.4m. 65° to C.A.		73052	13.0	14.0	1.00	0.014	S. All in of 0.047
15.6-19.23								15.6-19.23 ALTERED SILTSTONE - well layered w/ mod. Qtz veining parallel to layering - 45-50° to C.A. Local shearing accompanied by contorted veining. Graphite on shear slickensides. sulfide 1-3%		73053	14.0	14.63	0.63	0.048	
19.23-22.8								19.23-22.8 WEAK ZONE MATERIAL - intensely Qtz - carbonate albite veined area in sheared argillaceous siltstones. Local area 10-15cm across carry contort veins while other cores - cut core at various angles, sulfide content weak overall with 2-4cm wide zones carrying 5-10% crystalline py.		73054	14.63	15.6	0.97	0.129	
22.8-27.85								22.8-27.85 SHEARED ARGILLACEOUS SILTSTONE - range from thinly to thickly bedded dark charcoal grey. Veining weak with exception of localized 5-10 cm thick zones with contorted veining. GRANITIC FAULT ZONE from 26.6 to 27.2 m. weakly mineralized - sulfide <1%		73055	15.6	16.6	1.00	0.004	
27.85-28.3								27.85-28.3 BOULDER CONGLOMERATIC ARGILLITE		73056	16.6	17.6	1.00	0.008	
28.30-31.35								28.30-31.35 SHEARED ARGILLACEOUS SILTSTONE - dark charcoal grey shearing parallel to layering. Graphite slickenside		73057	17.6	18.6	1.00	0.002	
										73058	18.6	19.37	0.73	0.002	
										73059	19.37	19.87	0.50	0.032	
										73060	19.87	20.37	0.50	0.008	
										73061	20.37	21.0	0.67	0.010	
										73062	21.0	21.5	0.50	0.010	
										73063	21.5	22.0	0.50	0.004	
										73064	22.0	22.5	0.50	<0.002	
										73065	22.5	23.0	0.50	0.002	
										73066	22.8	24.0	1.20	0.002	
										73067	24.0	25.5	1.50	0.002	



LOCATION: M^c MASTER ZONE (Surface)

DIAMOND DRILL RECORD

PROJECT: M^c MASTER

HOLE NUMBER: 89 M-8

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1: 250	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz / ton
				CHLORITE	CALCITE	ALBITE						SILICA	from		
34.7	83		31						31.35 - 31.75	FAULT ZONE - graphitic slickensides.					
	100		32						31.75 - 32.0	SHEARED ARGILLACEOUS SILTSTONE - as from 28.3-31.35m					
33.22			33						32.0 - 33.07	ALTERED AND SHEARED BOULDER CONGLOMERATE					
	95		34						33.07 - 33.45	MEDIUM TO COARSE GRAINED LITHIC WACKE					
34.75			35						33.45 - 37.09	ALTERED AND SHEARED ARGILLACEOUS SILTSTONE - shearing parallel to layering ~ 65° to C.A. Intensely veined and faulted from 34.96 to 36.19 m. Graphite is abundant along fault slickensides. Minor sulfides					
36.27	85		36						37.09 - 40.84	ALTERED BOULDER CONGLOMERATIC ARGILLITE - becomes closely packed from 40.25 - 40.84 m. Moderate intensity Qtz - carb. veining with some silica flooding upper contact 50-55° to C.A. (obrupt) lower contact 60° to C.A. (obrupt). Sulfide content ranges from 0 to 5%.	73069	37.09	37.5	0.41	< 0.002
	100		37								73070	37.5	38.0	0.50	< 0.002
37.8			38								73071	38.0	38.5	0.50	< 0.002
	100		39								73072	38.5	39.0	0.50	< 0.002
39.31			40								73073	39.0	39.5	0.50	< 0.002
	96		41								73074	39.5	40.0	0.50	< 0.002
40.49			42						40.84 - 44.26	ALTERED ARGILLACEOUS SILTSTONE - weak to moderate Qtz - carbonate veining throughout. Sheared. Sulfide mineralization primarily < 1% several localized 1-2cm thick areas with up to 5% py. From 43 to 44.26 m veining decreases dramatically. Well layered ~ 75° to C.A.	73075	40.0	40.84	0.84	< 0.002
	100		43								73076	40.84	42.0	1.16	0.006
42.57			44								73077	42.0	43.5	1.50	< 0.002
43.89	95		45						44.26 - 45.41	FINE GRAINED LITHIC WACKE - dark grey - massive. weakly veined. some ARGILLACEOUS SILTSTONE AT 45.33					
46.4	101		46												
E. O. H															



LOCATION: M^cMASTER ZONE (SURFACE)

DIAMOND DRILL RECORD

PROJECT: M^cMASTER

HOLE NUMBER: 89-M-9

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	CALCITE	SILICA						from	to		
31.69	96	4	31						32.80-32.69 : ALTERED SILTSTONE: well bedded, 60° to c.A. minor greywacke, some bleaching	73432	29.80	31.00	1.20	<0.003	
			32							73433	31.00	32.00	1.00	20.003	
			33							73434	32.00	33.00	1.00	<0.003	
34.74	100		34						32.69-53.88 : TURBIDITE SEQUENCE: graded beds ranging from pebble conglomerate - lithic wacke to fine siltstone	73435	33.00	34.00	1.00	<0.003	
			35						close-packed coarse lithicwacke 35.76-36.21	73436	34.00	35.00	1.00	<0.003	
			36							73437	35.00	36.00	1.00	0.008	
37.18	84		37						Fault at 37.18 graphite on slickensides, colour change to light green @ 50° to c.A.	73438	36.00	37.18	1.18	<0.003	
			38						distinctly unmineralized.	- NOT SAMPLED -					
			39												
			40												
41.15	85		41												
			42												
42.97	101		43						short, slightly mineralized sections: 42.81-42.86; also 41.56-41.68.						
			44												
			45												
45.72	93		46												
			47												
			48												
48.77	99		49												
			50												
51.20	101		51												
			52												
			53												
53.44	97		54						53.88-85.67 ARGILLACEOUS SILTSTONE: dark grey, well bedded.						
			55												
			56												
			57												
			58												
59.13	96		59												
			60						60.15 to 73 m. intense gtz veining and silica flooding, vuggy gtz in open fractures at - 66.75 to 66.9 m, 67.10 to 67.28 and 67.70 to 67.77 m. Siltstone is brecciated in these areas and takes on a mylonitic appearance. Pyrite 5-10% throughout.	73439	61.66	62.50	0.84	0.012	
			61						Mineralization occurs along veins, fractures and as dissemination.	73440		63.00	0.50	0.020	
			62						Carbonate alteration is intense in areas of intense gtz veining. Weak albite alt? occurs sporadically within some gtz carbonate veins. Thin sections of LITHIC WACKE	73441		64.00	1.00	0.008	
			63							73442		65.00	1.00	20.003	
65.53	90		64							73443		66.00	1.00	<0.003	
			65							73444		67.00	1.00	0.018	
			66						from 71.1 to 71.3, 72.29 to 72.9 and from 74.27 to 74.58	73445		68.00	1.00	0.010	
67.05	96		67							73446		69.00	1.00	0.012	
			68												
			69												
			70							73447		70.00	1.00	0.022	

MISTAKE IN DRILLER'S BLOCKS

LITHIC WACKE



LOCATION: McMASTER ZONE (SURFACE)

DIAMOND DRILL RECORD

PROJECT: McMASTER

HOLE NUMBER: 89-M-9

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHALOPTE	CALCITE	ALBITE						From	to		
70.70	96	10	71						53.88 - 85.67 ARGILLACEOUS SILTSTONE cont'd.	73448	70.00	71.00	1.00	0.016	
			72						FAULTING occurs between 70.2 and 71 m with nubby core and	73449		72.00	1.00	0.014	
73.15	80		73						core loss. From 73 to 82.41 m poorly mineralized and weakly	73450		73.00	1.00	0.014	
			74						altered material occurs. Bedding and veining ~ 70° to C.A.	73460		74.00	1.00	0.004	
76.20	100		75												
			76												
			77												
			78												
79.25	96		79												
			80												
			81						81.6 to 81.8 m FAULTING	73451	80.21	81.00	0.79	0.014	
82.29	92		82						82.41 - 84.41 m - Qtz veining and silica flooding is more intense w/ 5-15%	73456	81.00	82.41	1.41	0.008	
			83						pyrite.	73452	82.41	82.90	0.49	0.010	
			84						84.41 to 84.50 m FAULTING.	73453	82.90	83.41	0.51	0.018	
85.34	91		85						85.67 to 84.79 TURBIDITE SEQUENCE	73454	82.41	84.41	1.00	0.090	
			86						Pale grey, green colour, greyish to charcoal grey argillaceous	73455	84.41	85.18	0.77	0.014	
			87						siltstone sections, black wispy mineral filling fractures (min.?)						
88.59	80		88												
			89												
			90												
91.44	102		91												
			92												
			93												
94.49	99		94						94.49 - 95.41 LAMINATED LITHIC WACKÉ pale grey, green colour.						
95.70	97		95						flattened clasts of siltstone & argillite lie in sheared foliation						
			96						planes ~ 60° to C.A. Unit grades fines upwards. Minor matrix						
			97						carbonate. No silica. minor albite.						
			98						95.41 - 98.41 SILTSTONE Bedding 70° to C.A. Grey, green occ. Qtz	73457	97.41	98.41	1.00	<0.003	
98.75	93		99						carb veinlet. No matrix carb. Silica & Albite, minor chl. - 3% py.	73458	98.41	99.8	1.39	0.008	
			100						dissem. on foliation planes & occasionally on margins of Qtz	73459	99.8	100.8	1.00	<0.003	
			101						veins & stringers.						
101.19	94		102						99.8 to 101.84 ALTERED SILTSTONE - dark grey to black inter laminated siltstone						
			103						in Qtz / albite flooded and veined (25% of rock). weak to Carb, silica						
			104						& Albite. weak chl. - 3% py.						
104.24	99		105						Coarse LITHIC WACKÉ (CONCL.?) pale green colour, flattened clasts of						
			106						arg. & siltstone chl. with						
106.37	80		107						101.84 to 105.78 ARGILLACEOUS SILTSTONE Foliated 55° to 60° to C.A.						
			108						105.78 to 106.37 TURBIDITE. grey green - minor carb, Qtz veining						

E.O.H.



LOCATION: (LEVEL): <u>McMASTER ZONE (SURFACE)</u>		DIAMOND DRILL RECORD		PROJECT: <u>McMASTER</u>	HOLE NUMBER: <u>89-M-10</u>
DIP: <u>-45°</u> DIRECTION <u>225° (TRUE)</u> SECTION <u>M-9, M-5</u>		CLAIM NUMBER:			
LATITUDE: <u>N</u>	LENGTH: <u>44.19 m (145 ft)</u>	ELEVATION:	LOCATION: <u>IDAHO CORE SHACK</u>		
DEPARTURE: <u>E</u>	CORE SIZE: <u>BQ</u>	DATE LOGGED: <u>Oct 29, 1989</u>	LOGGED BY: <u>J.S. WAIN, W.B.L.</u>		
STARTED: <u>Oct 28, 1989 D.S.</u>	FINISHED: <u>Oct 29, 1989 N.S.</u>	FINISHED: <u>Oct 28, 1989 D.S.</u>	SAMPLED BY: <u>W.B.L., W.A.H.</u>		
O.B. THICKNESS: <u>2.28 m</u>	STARTED: <u>Oct 28, 1989 D.S.</u>	FINISHED: <u>Oct 29, 1989 N.S.</u>	TOTAL RECOVERY: %		
B.R. THICKNESS:	STARTED: <u>Oct 28, 1989 D.S.</u>	FINISHED: <u>Oct 29, 1989 N.S.</u>	SURVEY: DEPTH <u>145' (44.19m)</u> BEARING <u>225° T</u> ANGLE Reading <u>-49.5°</u> Correc <u>-41°</u>		
CONTRACTOR: <u>F. BOIVENEAU</u>	CORE STORED: <u>IDAHO CORE SHACK</u>				

Night shift Oct 28-29 lost due to frozen pump and bent head.

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250 METERS	ALTERATION				FRACTURING	MINERAL %	GEOLOGY	PURPOSE: To test 2 ZONES above holes M-4 & M-5 COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	CALCITE	ALBITE	SILICA						from	to		
0-1.52			1							0 to 1.52 OVER BURDEN NO CORE.						
1.52-2.28			2							1.52 to 2.28 Rubby rock - possibly from bedrock as caving in fault zone	73461	2.28	3.28	1.00	0.174	
2.28-4.82			3							2.28 to 4.82 solid bedrock at 2.28 m. Casing to 2.74 m	73462	3.28	3.84	0.56	0.104	
4.82-6.40			4							ZONE MATERIAL - Light grey, chloritic, overall brecciated appearance. Sparse lenses of quartz and albite.	73463	3.84	4.18	0.34	0.034	
6.40-8.71			5							4.82-6.40 MINERALIZED AND ALTERED SILTSTONE: layered. 5.47-6.05 well mineralized zone material	73464	4.18	5.18	1.00	0.160	
8.71-10.60			6							6.40-15.45 CONGLOMERATIC ARGILLITE AND ARGILLACEOUS SILTSTONE: upper section altered, highly veined 10.60-13.6	73465	5.18	5.50	0.32	0.016	
10.60-12.9			7								73466	5.50	6.05	0.55	0.163	
12.9-14.70			8								73467	6.05	6.60	0.55	0.031	
14.70-16.27			9								73468	6.60	7.60	1.00	0.006	
16.27-18.60			10								73469	7.60	8.60	1.00	0.004	
18.60-20.71			11								73470	8.60	9.60	1.00	0.008	
20.71-23.04			12								73471	9.60	10.60	1.00	0.014	
23.04-24.69			13								73472	10.60	11.60	1.00	0.018	
24.69-25.71			14								73473	11.60	12.9	0.80	0.006	
25.71-27.1			15								73474	12.9	12.9	0.50	0.003	
27.1-28.71			16								73475	12.9	14.0	1.10	0.004	
28.71-30.11			17								73476	14.0	15.45	1.45	0.003	
30.11-31.6			18								73477	15.45	16.27	0.82	0.013	
31.6-33.04			19								73478	16.27	17.0	0.73	0.003	
33.04-34.69			20								73479	17.0	18.6	1.60	0.003	
34.69-35.71			21								73480	18.6	19.85	1.25	0.008	
35.71-37.1			22								73481	19.85	20.78	0.93	0.004	
37.1-38.71			23								73482	20.78	22.0	1.22	0.012	
38.71-40.11			24								73483	22.0	23.0	1.00	0.026	
40.11-41.6			25								73484	23.0	23.6	0.60	0.024	
41.6-43.04			26								73485	23.6	24.6	1.00	0.003	
43.04-44.69			27								73486	24.6	26.0	1.40	0.003	
44.69-46.27			28								73487	26.0	27.1	1.10	0.003	
46.27-47.81			29								73488	27.1	28.0	0.90	0.020	
47.81-49.27			30								73489	28.0	29.0	1.00	0.006	
49.27-50.71			31								73490	29.0	30.2	1.20	0.003	



LOCATION: Mc MASTER ZONE (SURFACE)

DIAMOND DRILL RECORD

PROJECT: Mc MASTER

HOLE NUMBER: 89 M - 10

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250	ALTERATION				MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton			
				CHLORITE	CALCITE	ALBITE	SILICA					from	to					
30.90	100	6	31						23.60 - 35.80 ARGILLACEOUS SILTSTONE: dark grey, well layered	73491	30.2	31.0	0.80	<0.003				
32.0	98		32						Traces of pyrite @ 33.01-33.56, graphitic slickensides at 33.56.	73492	31.0	32.0	1.00	0.006				
33.53	98		33						chloritic greywacke section 34.81-35.55... minor pyrite 35.50-35.55	73493	32.0	33.0	1.00	0.012				
35.05	98		34							73494	33.0	33.8	0.80	0.008				
36.57	95		35						35.80 - 37.30 CONGLOMERATIC SECTION: mainly argillaceous but also altered and bleached.	73495	33.8	35.5	1.70	<0.003				
38.1	107		36						37.30 - 39.90 - WEAK ZONE MATERIAL - calcareous, slightly pyritic	73496	35.5	37.0	1.50	<0.003				
41.15	89		37						39.90 - 44.19 (EOH) CHLORITIC GREYWACKE: uniform dark green	73497	37.0	38.0	1.00	0.004				
42.87	98		38						Not sheared, aside from calcite lenses, very sparse grey pyrite - 40.95-41.08.	73498	38.0	39.0	1.00	0.012				
44.7			39						Very chloritic 43.20-43.30	73499	39.0	40.0	1.00	<0.003				
			40							73500	40.0	41.0	1.00	<0.003				
			41															
			42															
			43															
			44															
									END OF HOLE - 44.19 meters (145 feet)									



LOCATION '(LEVEL)' M ^c MASTER ZONE (SURFACE)		DIAMOND DRILL RECORD		PROJECT: M ^c MASTER	HOLE NUMBER: 89 M-11
DIP: -60° Az 225 true Section M-4, M-5				CLAIM NUMBER:	
LATITUDE: N	LENGTH: 72.23m (237')	ELEVATION:			
DEPARTURE: E	CORE SIZE: BQ	DATE LOGGED: Oct 31, Nov. 1, 1989	LOCATION: IDAHO CORE SHACK		
STARTED: Oct 29, 1989 NS	FINISHED: Oct 30-31, 1989 N.S.	LOGGED BY: J.T.S., W.B.L.	SAMPLED BY: J.T.S. W.B.L.		
O.B. THICKNESS: 2.44 meters	STARTED: Oct 29, 1989 NS	FINISHED: Oct 29, 1989 NS	CASING: 3.05m		
B.R. THICKNESS:	STARTED: Oct 30, 1989 D.S.	FINISHED: Oct 30-31, NS	TOTAL RECOVERY: %		
CONTRACTOR: F. B DIVENEAU	CORE STORED: IDAHO CORE SHACK		SURVEY: DEPTH: 72.23m (237')		
			BEARING: 225 (T)		
			Reading: -63.5°		
			Angle: -56°		

DRILLING INTERVAL	% CORE RECOVERED	BOX NUMBER	SCALE 1:350 METERS	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	ALBITE	SILICA						FROM	TO		
			1						0-2.44 OVERBURDEN No CORE (RUBBLE 1.67-2.44)						
2.44-3.05	60	2.44	2					ZM	2.44-5.58 CHLORITIC ZONE MATERIAL: Mineralized pyrite		73536	2.44	3.0	0.56	0.139
3.05-3.65	110		3								73537	3.0	3.50	0.50	0.104
			4								73538	3.50	4.00	0.50	0.086
			5								73539	4.0	4.50	0.50	0.029
			6								73540	4.50	5.0	0.50	0.035
			7								73541	5.0	5.58	0.58	0.056
6.70	96		8						5.58-6.76 ALTERED SILTSTONE		73542	5.58	6.00	0.72	0.012
			9								73543	6.0	6.76	0.76	0.016
			10					ZM	6.76-8.73 CHLORITIC ZONE MATERIAL		73544	6.76	7.26	0.50	0.008
9.75	100		11								73545	7.26	7.76	0.50	0.026
			12								73546	7.76	8.26	0.50	0.124
			13								73547	8.26	8.73	0.47	0.193
			14						8.73-11.38 CONGLOMERATIC ARGILLITE		73548	8.73	9.73	1.00	0.012
12.80	93	9.42	15								73549	9.73	11.38	1.65	0.018
			16						11.38-22.98 SHEARED SILTSTONE: well layered, entire interval cut by many quartz lenses, short highly contorted sections bedding ~ 50° to 0 C.A. Veining dominantly parallel and/or subparallel to layering. Sulphide content less than 2%. Graphite developed along some bedding planes.		73550	11.38	12.0	0.62	0.016
15.85	103	16.50	17								73551	12.0	13.0	1.00	0.028
18.89	98		18								73552	13.0	14.0	1.00	0.016
			19								73553	14.0	15.0	1.00	0.008
			20								73554	15.0	16.0	1.00	0.012
			21								73555	16.0	17.0	1.00	0.010
21.95	100		22								73556	17.0	18.0	1.00	10.003
			23								73557	18.0	19.0	1.00	0.006
			24								73558	19.0	20.5	1.50	10.003
24.99	106	23.62	25						22.98-23.62 MASSIVE GREYWACKE - grey, weakly veined, sulphide < 1% 23.62-28.76 GRAPHITIC FAULT ZONE 75° to C.A. 23.76-29.02 HIGHLY ALTERED CONGLOMERATIC ARGILLITE: Brecciated appearance. Qtz-carbonate veining is intense and is contorted. Sulphide 1-5% w/ py. & minor arsenopyrite?		73559	20.5	22.0	1.50	0.004
			26								73560	22.0	22.98	0.98	0.006
28.04	89		27								73561	22.98	23.62	0.64	10.003
			28								73562	23.62	24.0	0.38	0.010
31.0	102	30.80	29								73563	24.0	25.0	1.00	10.003
			30								73564	25.0	25.5	0.50	10.003
											73565	25.5	26.0	0.50	0.010
											73566	26.0	26.5	0.50	0.004
											73567	26.5	27.0	0.50	0.006
											73568	27.0	28.5	0.50	0.012
											73569	28.5	29.0	0.50	0.022
											73570	29.0	28.5	0.50	0.010
											73571	28.5	29.02	0.52	0.010
											73572	29.02	70.5	1.48	0.008
									29.02 - WEAKLY SHEARED ARGILLACEOUS SILTSTONE:						



DIAMOND DRILL RECORD

LOCATION: _____

PROJECT: _____

HOLE NUMBER: **M-11**

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250	ALTERATION		FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/tm			
											FROM	TO					
31									29.02-38.61 WEAKLY SHEARED ARGILLACEOUS SILTSTONE:	73573	30.5	32.0	1.50	0.009			
32									29.8 to 30 m. Graphitic fault zone. Qtz-carbonate veining less intense	73574	32.0	33.50	1.50	0.006			
33									is primarily parallel to subparallel to layering. Small local 5-10cm	73575	33.50	35.0	1.50	0.003			
34									zones contain contorted veining. Sulphides < 2%	73576	35.0	36.5	1.50	0.008			
35										73577	36.5	38.61	2.11	0.010			
36										73578	38.61	40.11	1.50	0.009			
37									38.61-42.70 ALTERED CONGLOMERATIC ARGILLITE:	73579	40.11	41.61	1.50	0.002			
38									Noticeable increase in sulphides including < 1% arsenopyrite?	73580	41.61	42.70	1.09	0.009			
39									Qtz-carbonate veining is not intense other than 5-10cm localized	73581	42.70	43.70	1.00	0.002			
40									clots.	73582	43.70	44.69	0.99	0.009			
41									42.70-44.69 CHLORITIC GREYWACKE: sparsely disseminated pyrite	73583	44.69	45.19	0.50	0.019			
42									massive	73584	45.19	46.69	0.50	0.022			
43									44.69-47.09 ZONE MATERIAL: light grey quartz-albite-carbonate	73585	45.69	46.19	0.50	0.018			
44									sharp contact - mainly vein material. Sulphides (py, arsenopy) 5-8% rock.	73586	46.19	46.69	0.50	0.047			
45									as disseminations and along fractures and in veinlets	73587	46.69	47.09	0.40	0.040			
46									47.09-57.52 WEAKLY CHLORITIC GREYWACKE:	73588	47.09	48.09	1.00	0.012			
47									short altered section near top of interval. Very massive								
48									appearance. Mineralization is very weak to nil. Veining is also								
49									noticeably absent except for minor ones that appear to vva								
50									parallel to layering								
51																	
52																	
53																	
54									SLIGHTLY MORE ALTERED starting 55.47 and below.								
55																	
56																	
57									57.52-72.29 ALTERED ARGILLACEOUS SILTSTONE	73589	57.9	56.9	1.50	0.002			
58									EOH.	73590	56.9	57.52	0.62	0.006			
59									finely pyritic 57.52-59.30. Pyrite is disseminated along	73591	57.52	58.0	0.48	0.002			
60									qtz-carbonate altered plane parallel to compositional layering.	73592	58.0	58.5	0.50	0.006			
61									Sulphide content range from 2-10%	73593	58.5	59.3	0.80	0.009			
62										73594	59.3	60.8	1.50	0.002			
63										73595	60.8	61.8	1.00	0.002			
64																	
65																	
66																	
67																	
68																	
69																	
70																	

3.40m of 0.23 oz/tm Au



LOCATION (LEVEL): McMASTER ZONE (SURFACE)		DIAMOND DRILL RECORD		PROJECT: McMASTER R	HOLE NUMBER: 89 M-12
DIP: -40° DIRECTION 225° TRUE		LATITUDE: N LENGTH: 5700m (187feet) ELEVATION:		CLAIM NUMBER: CARD #3 FR.	
DEPARTURE: E		CORE SIZE: BQ DATE LOGGED: Nov 1, 1989		LOCATION: IDAHO CORE SWACK	
STARTED: Oct 31, 1989 D.S.		FINISHED: Oct 31, Nov 1, 1989 N.S. LOGGED BY: J.T.S., WBL		SAMPLED BY: J.T.S., WBL, SLS	
O.B. THICKNESS: 5.49m (18ft)		STARTED: OCTOBER 31 1989 D.S. FINISHED: OCTOBER 31 1989 D.S. CASING: 6.71m		TOTAL RECOVERY: %	
B.R. THICKNESS:		STARTED: October 31, 1989 D.S. FINISHED: OCTOBER 31, N.S.		SURVEY: ACID ANGLE	
CONTRACTOR: F. BOIVENEAM		CORE STORED: IDAHO MINE		DEPTH: 57.00m BEARING: 225°(6) Reading: -47° Correct: -38°	

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE METERS	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	CALCITE	ALBITE						SILICA	FROM		
			1						0-5.49 OVERBURDEN: NO CORE						
			2						minor rubble 4.88-5.49, SLIGHTLY ALTERED SILTSTONE						
			3												
			4												
			5												
5.49-6.71	82	1	6						5.49-6.41 SLIGHTLY ALTERED SILTSTONE:		73501	5.49	6.41	0.92	20.003
			7						6.41-15.00 WELL MINERALIZED ZONE MATERIAL: high pyrite, pyrrhotite-arsenopyrite content, typical of higher grade quartz-albite carbonate alteration, sulfide content 15-20%, at top veining 30% to CA.		73502	6.41	7.00	0.59	0.009
			8						Rubblly upper contact		73503	7.00	7.80	0.80	0.100
			9						veining slightly convoluted at 11.95-12.38. veining 70° to CA at 12.80.		73504	7.80	8.00	0.50	0.056
			10						graphitic along fractures at 13.05.		73505	8.00	8.50	0.50	0.024
			11						sheared, abrupt lower contact		73506	8.50	9.00	0.50	0.164
			12						15.00-17.63 GRAPHITIC ARGILLACEOUS SILTSTONE: MAJOR FAULT ZONE 15.85-16.70 intense graphite development faulted, lower contact - graphitic... Possible andesite dyke - 17.05-17.52		73507	9.00	9.50	0.50	0.188
			13						17.63-23.97 CHLORITIC GREYWACKE: Dark, relatively uniform, massive - not sheared.		73508	9.50	10.50	1.00	0.198
			14								73509	10.50	11.00	0.50	0.092
			15								73510	11.00	11.50	0.50	0.107
			16								73511	11.50	12.00	0.50	0.087
			17								73512	12.00	12.50	0.50	0.050
			18								73513	12.50	13.00	0.50	0.175
			19								73514	13.00	13.50	0.50	0.264
			20								73515	13.50	14.00	0.50	0.196
			21								73516	14.00	14.50	0.50	0.182
			22								73517	14.50	15.00	0.50	0.188
			23								73518	15.00	15.50	0.50	0.012
			24								73519	15.50	16.00	0.50	0.003
			25								73520	16.00	17.00	1.00	0.011
			26								73521	17.00	18.00	1.00	20.003
			27								73522	18.00	19.00	1.00	20.006
			28												
			29												
			30												

8.59m = 28.2 ft

weighted average
15.2 oz/ton
over
6.5m = 21.32 feet



LOCATION: McMASTER SURFACE

DIAMOND DRILL RECORD

PROJECT: McMASTER

HOLE NUMBER: 89 M-12

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
												FROM	TO		
31.70	72	4	31						31.05-42.43 CONGLOMERATIC ARGILLITE: many boulders Bedding 80-90° to core axis. Intense graphite throughout in narrow sections for example 35.24. Large rounded clasts common. entire section slightly altered, short silty sections such as 36.76-36.93 occur.	73523	31.05	31.75	0.70	<0.003	
34.13	107	339	32							73524	33.50	34.30	0.80	0.008	
			33							73525	34.50	35.50	1.00	<0.003	
			34							73526	35.50	36.50	1.00	<0.003	
			35												
			36												
			37												
37.49	89		38												
39.01	86	40.23	39												
40.23	100		40												
41.75	89		41						42.43-46.61 ARGILLACEOUS SILTSTONE: dark grey						
43.28	95		42												
44.80	87		43												
46.33	95	47.4	44						46.61-50.54 CONGLOMERATIC ARGILLITE: mainly small pebbles						
47.85	96		45												
49.37	100		46												
50.90	98		47						50.54-54.96 ALTERED SILTSTONE (MINERALIZED) sparse pyrite in quartz-albite-carbonate altered zone.	73527	50.34	51.00	0.66	0.010	
52.42	104	54.25	48							73528	51.00	51.50	0.50	0.010	
53.95	93		49							73529	51.50	52.00	0.50	0.072	
55.41	98		50						54.96-57.00 (EOH) ALTERED SILTSTONE - less mineralized thin bedded.	73530	52.00	52.50	0.50	0.081	
57.00	98		51							73531	52.50	53.00	0.50	0.050	
			52							73532	53.00	54.00	1.00	0.032	
			53							73533	54.00	55.00	1.00	0.020	
			54							73534	55.00	56.00	1.00	0.004	
			55							73535	56.00	57.00	1.00	<0.003	
			56												
			57												

END OF HOLE 57.00 meters
(187 feet)

2.50 ~ 4.0053
oz/ton Au



LOCATION (LEVEL): M ^c MASTER ZONE (SURFACE)		DIAMOND DRILL RECORD		PROJECT: M ^c MASTER	HOLE NUMBER: 89 M-13
DIP: -40° Az. 225°				CLAIM NUMBER:	
LATITUDE: N	LENGTH: 92.05 m (302ft)	ELEVATION:	LOCATION: IDAHO CORE SHACK		
DEPARTURE: E	CORE SIZE: BQ	DATE LOGGED: Nov. 5, 1989	LOGGED BY: W.B.L., J.T.S.		
STARTED: Nov. 1, 1989	FINISHED: Nov. 2, 1989	LOGGED BY: W.B.L., J.T.S.	SAMPLED BY: W.B.L.		
O.B. THICKNESS: 0.61 m	STARTED: Nov. 1, 1989 D.S.	FINISHED: Nov. 1, 1989 D.S.	CASING:		
B.R. THICKNESS: 92.05 m	STARTED: Nov. 1, 1989 D.S.	FINISHED: Nov. 2, 1989 D.S.	TOTAL RECOVERY: %	SURVEY: DEPTH BEARING Reading Correct	
CONTRACTOR: F. BOISEVENU	CORE STORED: IDAHO CORE SHACK		45.72 m	225°	+45° -37°
			92.05 m	"	-45° -37°

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: To test extension of Zone Material located in DDH COMMENT: m-9 to the south.	SAMPLE NUMBER	METERS		Au oz/ton	
				CHLORITE	ALBITE	SILICA						from	to		LENGTH METERS
0.91			1						0. - 0.61 OVERBURDEN 0.61 - 0.91 RUBBLE CASING TO 1.52 m		73078	0.91	1.50	0.57	0.139
	43		2						0.91 - 3.66 ZONE MATERIAL - Near surface weathering - very broken up & rusty core. Intensely brecciated, gtz - calcite - albite flooding. Sulfide content is high ~ 15% mainly pyrite with oipy in localized patches.		73079	1.50	2.00	0.50	0.185
3.66			3								73080	2.00	2.50	0.50	0.066
	96		4						3.66 - 9.42 ARGILLACEOUS SILTSTONE - strongly sheared shearing has imparted a strong foliation (parallel to layering?) that runs 5-15° to C.A. gtz - carbonate alteration moderate to strong as fine hairline coatings along foliation planes. Pyrrhotite is main sulfide to 15% - strongly magnetic - ~ 5% py - 1-2% asp.		73081	2.50	3.00	0.50	0.104
6.70			5								73082	3.00	3.66	0.66	0.062
	98		6								73083	3.66	4.00	0.34	0.260
9.15			7								73084	4.00	4.50	0.50	0.029
	98		8								73085	4.50	5.00	0.50	0.208
12.8			9								73086	5.00	5.50	0.50	0.049
	101		10								73087	5.50	6.00	0.50	0.032
15.85			11								73088	6.00	6.50	0.50	0.026
	94		12								73089	6.50	7.00	0.50	0.074
18.82			13								73090	7.00	7.50	0.50	0.014
	99		14								73091	7.50	8.00	0.50	0.026
21.99			15								73092	8.00	8.50	0.50	0.092
	100		16								73093	8.50	9.00	0.50	0.068
24.77			17								73094	9.00	9.42	0.42	0.030
	99		18								73095	9.42	10.00	0.58	0.010
28.09			19								73096	10.00	10.50	0.50	0.024
	99		20								73097	10.50	11.00	0.50	0.062
			21								73098	11.00	11.50	0.50	0.042
			22								73099	11.50	12.00	0.50	0.026
			23								73100	12.00	12.50	0.50	0.026
			24								73101	12.50	13.00	0.50	0.167
			25								73102	13.00	13.50	0.50	0.208
			26								73103	13.50	14.00	0.50	0.158
			27								73104	14.00	14.34	0.34	0.042
			28								73105	14.34	15.00	0.66	0.018
			29								73106	15.00	16.00	1.00	0.004
			30								73107	16.00	17.00	1.00	0.006
											73108	17.00	18.00	1.00	0.008
											73109	18.00	19.00	1.00	0.016
											73110	19.00	20.00	1.00	0.006
											73111	20.00	21.00	1.00	0.038
											73112	21.00	21.90	0.90	0.018
											73113	21.90	22.40	0.50	0.044
											73114	22.40	23.00	0.60	0.008
											73115	23.00	24.00	1.00	0.024
											73116	24.00	25.00	1.00	0.004
											73117	25.00	25.60	0.60	0.016
											73118	25.60	26.00	0.40	0.038
											73119	26.00	26.50	0.50	0.046
											73120	26.50	27.00	0.50	0.078
											73121	27.00	27.40	0.40	0.010
											73122	27.40	28.00	0.60	0.072
											73123	28.00	29.00	1.00	0.008
											73124	29.00	30.28	1.28	0.006



LOCATION: Mc MASTER ZONE

DIAMOND DRILL RECORD

PROJECT:

Mc MASTER

HOLE NUMBER:

87 M-13

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1: 250	ALTERATION				FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au oz/ton
				CHLORITE	CALCITE	ALBITE	SILICA						from	to		
70.71	100	10	71							69.72 - 73 ARGILLACEOUS SILTSTONE - as from 58.77 - 69.0 slight increase in gta - carbonate veining. Sulfides ~ 1%						
72.24	100		72													
73.76	105	11	73							73 - 76.81 CONGLOMERATIC ARGILLITE - to argillaceous lithic wacke. Dark charcoal grey. Thin interbeds within main Argillaceous siltstone unit. At 74 m. grades to a very fine grained argillaceous siltstone.						
75.28	93		74													
76.81	95	95	75													
78.33	95		76													
79.84	105	79.33	77							76.81 - 92.05 (E.O.H) TURBIDITE SEQUENCE - medium green-grey moderately to strongly chlorite altered. Massive unit. At 76.9m small fault zone						
81.38	100		78	X												
82.29	105	94	79	X						76.81 - 86.37 - Grey wacke						
			80								86.37 - 86.9m Fine grained lithic wacke					
		94	81	Y						86.9 - 87.72m Grey wacke						
			82								87.72 - 89 Medium to coarse lithic wacke					
		94	83	Y						89.0 - 90.5 Intensely sheared and altered lithic wacke						
			84								90.5 - 92.05 Medium grained lithic wacke - massive, poorly veined, light grey greenish. Clasts aligned 80° to C.A. minor sulfides.					
		94	85	Y												
			86													
		94	87	Y												
			88													
		94	89	Y												
			90													
		94	91	Y												
			92													
		94	93													
			94								E.O.H.					

1975 DIAMOND DRILL HOLES M-1 TO M-7

RELOGGED 1989



DIAMOND DRILL RECORD

LOCATION '(LEVEL)': SURFACE M^cMASTER PROJECT: M^cMASTER HOLE NUMBER: M-1

DIP: -37° BEARING: 225° T

LATITUDE: 4291.6 N LENGTH: 49.7 m ELEVATION: 1476.7 CLAIM NUMBER: CARO #3 FRAC.

DEPARTURE: 2306.4 E CORE SIZE: BQWL DATE LOGGED: Relogged sept 28/69 LOCATION:

STARTED: Oct 6 1975 FINISHED: Oct 7 1975 LOGGED BY: D.J.G., JTS 1989 SAMPLED BY:

O.B. THICKNESS: 3.66 m. STARTED: FINISHED: CASING:

B.R. THICKNESS: STARTED: FINISHED: TOTAL RECOVERY: %

CONTRACTOR: Shepard Enterprises Ltd CORE STORED: at M^cMaster Camp. 1989.

SURVEY:		ANGLE	
DEPTH	BEARING	Reading	Correc
SURFACE	225 T	-37	

INTERVAL	% CORE RECOVERED	BOX NUMBER	SCALE METERS	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne oz/ton.
										to	from		
1			1				1	0-3.66m OVERBURDEN					
2			2				2						
3			3				3						
4		3.66	4			CHLORITIC GREYWACKE	4	3.66-16.80 [QUARTZ-ALBITE ROCK] 3.66-11.80 CHLORITIC GREYWACKE. slightly mineralized with py + po, minor quartz albite carbonate alteration	4-1	3.66	4.16	0.50	0.020
5			5				5		-3	4.80	5.70	0.90	0.010
6			6				6		-4	5.70	6.30	0.65	0.080
7			7				7		-5	6.30	7.20	0.90	0.010
8			8				8		-6	7.20	8.30	1.10	0.010
9			9				9		-7	8.30	9.20	0.90	0.010
10			10				10		-8	9.20	10.65	1.45	0.020
11			11				11		-9	10.65	12.10	1.45	0.030
12			12				12		-10	12.10	12.65	0.55	0.005
13			13				13		-11	12.65	13.25	0.60	0.015
14			14				14		-12	13.25	14.13	0.88	0.015
15			15				15		-13	14.13	15.40	1.27	0.035
16			16				16		M-14	15.40	16.80	1.40	0.16
17			17				17		-15	16.80	17.70	0.90	0.010
18			18				18		-16	17.70	20.10	2.40	0.005
19			19				19		-17	20.10	21.75	1.65	±rc
20			20				20		-18	21.75	23.35	1.60	±rc
21			21				21		-19	23.35	25.50	2.15	±rc
22			22				22		-20	25.50	26.08	0.81	0.065
23			23				23		-21	26.08	27.49	0.50	0.023
24			24				24		-22	27.49	29.15	1.65	0.010
25			25				25		-23	29.15	30.64	1.10	0.020
26			26				26						
27			27				27						
28			28				28						
29			29				29						
30			30				30						

NOTE: Many of the wooden blocks have faded and can not be read (approx in pencil).

bed 20 inc 7% CLV 7% QZ PAR CLV
 1% CA, 2.6% (0.1%-7.0%) PDR
 CLV 10. 1% CA, 2.6% (0.1%-7.0%) PDR
 3.0% (0.0%-15%) PY 0.1% MSCP

→ data interval 1m



DIAMOND DRILL RECORD

LOCATION: SURFACE McMASTER ZONE

PROJECT: McMASTER

HOLE NUMBER: M-1

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1: 250 METERS	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne oz/ton			
										to	from					
30.7	30.7	30.7	31				31	30.70 - 36.6		30.64	31.20	0.56	Trc			
			32				32			31.20	32.01	0.81	0.005			
			33				33			31.61	32.51	0.90	0.025			
			34				34	minor quartz - calcite veining		32.51	33.60	1.09	0.010			
33.6			35				35			33.60	34.70	1.10	0.020			
			36				36			34.20	35.40	0.70	0.030			
36.6		36.6	37				37	36.6 - 46.9		35.40	35.75	0.35	0.020			
			38				38	greenish greywacke - lithicwacke.		35.75	37.55	1.80	0.035			
			39				39	sulfides most abundant at 38.0 in quartz stockworks		37.55	37.80	0.25	0.025			
39.6			40				40			37.80	39.05	1.25	0.020			
			41				41			39.05	39.53	0.48				
			42				42			39.53	41.14	1.61	0.040			
43.7		43.5	43				43			41.14	42.04	0.90	0.005			
			44				44	prominent greenish colour, minor lithic fragments to 5mm.		42.04	42.64	0.60	Trc			
			45				45			42.64	43.00	0.36	0.010			
46.7			46				46			43.00	44.52	1.52	Trc			
			47				47	46.9 - 48.2		44.52	46.20	1.68	0.005			
			48				48	LAMINATED SILTSTONE		46.20	47.40	1.20	Trc			
49.7		49.7	49				49	48.2 - 49.7		47.40	48.74	1.34	Trc			
			50				50	BLACK ARGILLITE.		48.74	49.70	0.96	Trc.			
								END OF HOLE 49.7 meters.								

qtz - carb
alt.



DIAMOND DRILL RECORD

PROJECT:

HOLE NUMBER: M-2

LOCATION:

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE 1:250 METERS	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
												to	from		
31.2	4		31						371 25.47-43.75m CONGLOMERATIC ARGILLITE			33.25	1.95	TRC	
	32.7		32									35.25	3.18	TRC	
			33									36.43			
			34						BOX MISSING BOX FOUND LATER						
34.3			35						ALL CONGLOMERATIC ARGILLITE, Bedding mostly 70-80° to c.a.			36.43	2.90	TRC	
			36									39.39			
			37												
37.5			38						very few pebbles toward end of interval - argillite			39.39	1.84	TRC	
	39.6		39						short quartz alteration zones 41.2			41.17			
40.5			40						Bedding @ 42.8 is 50° to c.a.			43.75	2.58	TRC	
			41												
			42												
43.5			43						43.75 - 44.55m QUARTZ ZONE - CONTACT ZONE			44.71	0.96	0.015	
			44						Rock greenish grey / silty sh. wacka - siltstone			45.55	0.84	0.005	
			45						44.55 - 45.55m QUARTZ ZONE (0.7% - 0.9%) by Petrographic graded sequences.			46.39	2.84	TRC	
	46.5		46						AS 0.2% PR. TRC (-TRC) CR. 0.9%						
46.6			47						(0% - 5.0%) PY			49.33	0.94	0.010	
			48									49.43	0.60	0.020	
47.6			49									50.68	0.75	0.005	
			50									52.57	0.89	TRC	
49.1			51												
50.5			52									52.87	3.30	TRC	
52.0	52.6		53												
			54												
53.0			55						very short quartz - albite weakly mineralized zones @ 55.5 10cm wide			57.05	1.18	TRC	
			56									57.05	3.50	TRC	
55.5			57												
57.5			58												
			59												
	59.4		60									61.00	0.45	0.005	
			61									61.50	0.50	TRC	
			62						BOX MISSING			62.70	1.40	0.005	
			63									62.52	3.62	TRC	
			64												
			65												
			66												
			67						END OF HOLE			67.92	1.40	TRC	
	67.6		68						END OF HOLE 68.19m						
	68.9		69									18.19	0.27	TRC	
			70												

Box 5 Missing

Box 9 Missing



DIAMOND DRILL RECORD

LOCATION: _____

PROJECT: _____

HOLE NUMBER: **M3**

DRILLING INTERVAL	% CORE RECOVERED	BOX NUMBER	SCALE	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE COMMENT	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
												to	from		
		2	31						31.0 - 35.90 PY Graphitic SILTSTONE. platy cleavage.	M3-10	30.96	30.16	0.80	trc	
		32.0	32						Pyrite layer at 32.75 FOLDED in an ellipse.	M3-11	32.37	30.96	1.41	0.020	
			33							M3-12	34.45	32.37	2.08	trc	
			34							M3-13	35.45	34.45	1.00	trc	
			35							M3-14	36.90	35.45	1.45	0.005	
			36						calcite-quartz zone 36.6-36.9, faulted, bedding at 55° to CA.	M3-15	39.20	36.90	2.30	trc	
			37							M3-16	40.24	39.20	1.04	trc	
			38							M3-17	41.10	40.24	0.86	trc	
			39												
			40												
			41												
			42												
			43						Fault breccia at 45.7.	M3-18	44.60	41.10	3.50	trc	
			44						Quartz veining over short intervals. Faulted 45.0-50m. brecciated + healed.	M3-19	45.80	44.60	1.20	0.005	
			45							M3-20	47.80	45.80	2.00	0.005	
			46												
			47												
			48												
			49												
			50						(50.90m to 62.10m) short TURBIDITIC ARGILLITE and Turbidite section	M3-21	50.30	47.80	2.50	trc	
			51						MUDBALL SLATE, GR ARGL; BED/CLV/GR 80°; 4% OZ PAR CLV 410°; 5% CA; 0.2% PR; 0.2% PY; 0% AS, CP; Lithic wacke at top of interval	M3-22	50.90	50.30	0.60	trc	
			52						Lost water 61.0m	M3-23	53.20	50.90	2.30	trc	
			53												
			54												
			55												
			56												
			57												
			58												
			59												
			60												
			61						(62.10 - 79.06m) TURBIDITIC ARGILLITE and Turbidite section	M3-27	61.32	59.10	2.22	trc	
			62						MUDBALL SLATE, GR ARGL, BED/CLV/CL 80°-100°; 5% OZ (1% - 20%) PAR CLV 410°; 5% CA; 0.2% PR; 0.2% PY; 0% AS, CP; Lithic wacke at top of interval	M3-28	62.10	61.32	0.78	trc	
			63												
			64												
			65						Graded cycles throughout from coarse lithic wacke to finely laminated silty-argillaceous tops.	M3-29	64.00	62.10	1.90	0.070	
			66												
			67												
			68												
			69												
			70						abundant carbonate veining throughout Box 8.	M3-30	66.00	64.00	2.00	trc	
			71												
			72												
			73												

? Blocks not readable
Box 3 broken but still useful

46.0
46.5

56.0

Box 7 MISSING

1.90 m of 0.070
0.02 trc



DIAMOND DRILL RECORD

LOCATION :

PROJECT :

HOLE NUMBER :

M 4

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE COMMENT	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
												to	from		
		4	51						19.5 - 39.62 CONGLOMERATIC ARGILLITE :						
		33.3	52								M4-19	33.81	30.50	3.31	trc
			53						Less Graphite		M4-20	35.75	33.81	1.94	0.005
			54								M4-21	37.47	35.75	1.72	trc
37.9	core loss		55								M4-22	38.52	37.47	1.05	trc
39.1			56								M4-23	39.62	38.52	1.10	0.005
39.62		40.1	57						39.62 - 49.10 TURBIDITE		M4-24	40.75	39.62	1.13	0.015
39.62			58						Interbedded, Wacke & Argill. Bed/CLU 50%; 3% BZ PAR CLU > 10% trc PR; 0.4% PY; 0% AS - CP		M4-15	42.38	40.75	1.63	trc
			59						Coarse lithic wacke to fine siltstone cycles.		M4-16	42.70	42.38	0.32	0.005
			60								M4-27	43.40	42.70	0.70	0.005
			61								M4-28	44.11	43.40	0.71	trc
			62								M4-29	45.33	44.11	1.22	trc
			63						Intense graphitic on slickensides 45.7, + gtz breccia		M4-30	46.00	45.33	2.67	trc
47.6		47.2	64						2cm wide graded beds at 47.7.		M4-31	49.10	48.00	1.10	trc
49.10		49.0	65						(49.10 - 67.50)		M4-32	50.59	49.10	1.49	trc
			66						Green Argill. Argill. CLU/CL 80%; 6% trc (trc-80%) GZ PAR CLU 80% 2% CH; 0.3% PR; trc CP; 0.4% PY; 0% AS;		M4-33	51.80	50.59	1.22	0.070
			67						Brecciated and altered 49.5, Quartz breccia zone 49.5-51.2.		M4-34	54.60	51.80	2.80	trc
			68						51.2 - sheared + brecciated TURBIDITE		M4-35	57.05	54.60	2.45	trc
			69								M4-36	58.35	57.05	1.30	0.005
			70								M4-37	60.23	58.35	1.88	trc
			71								M4-38	61.70	60.23	1.47	0.005
			72						TURBIDITE, green, coarse to fine cycles.		M4-39	62.85	61.70	1.15	trc
			73								M4-40	65.12	62.85	2.27	trc
			74								M4-41	67.50	65.12	2.38	trc
67.50		67.5	75						66.0 - ESH GRAPHITIC FAULT in conglomeratic argillite and chloritic greywacke.						



DIAMOND DRILL RECORD

PROJECT:

HOLE NUMBER: M-5

LOCATION:

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION			FRACTURING	MINERAL	GEOLOGY	PURPOSE COMMENT	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
												to	from		
31.8		2	31						30.20 - 35.90 CONGLOMERATIC ARGILLITE: chloritic ... pebbles common throughout	MS-5	31.45	30.43	102	trc	
			32							MS-6	32.08	31.45	143	trc	
			33							MS-7	34.57	32.08	149	trc	
34.5			34												
36.0		36.0	35						35.90 - 49.70 TURBIDITE: mainly fine grained silty cycles sheared in top of interval.	MS-8	36.33	34.37	196	trc	
37.0			36						2 wooden blocks at 37.0 - HOLE	MS-9	38.00	36.33	167	trc	
			37												
			38												
40.0			39						Fault convolutions at 40.3	MS-10	40.00	38.00	200	trc	
			40							MS-11	40.60	40.00	150	trc	
			41												
			42							MS-12	43.04	40.60	244	trc	
			43						graphitic slickensides 44.8, 10° to c.A. graphite on slickensides 85° to c.A.	MS-13	44.60	43.04	156	trc	
			44						GRAPHITIC	MS-14	46.35	44.60	125	trc	
			45							MS-15	48.25	46.35	190	trc	
49.70			46						(49.70 - 64.85 m) Brecciated Chert, Interbedded Wacke & Arg. Bed/CLV/CL/GF 50%, 17% (43-30%) QZ PAR CLV 2.0%; 32% CH; trc (0.90 - trc) AS; 0.2% PR; trc (0.90 - trc) CP; 0.3% (0.190 - 1.5%) PY.	MS-16	49.70	48.25	145	trc	
			47						MAJOR FAULT BRECCIA - intense from 49.70 - 55.0m	MS-17	52.47	49.70	277	0.010	
			48						Light grey silicification common (bleaching)	MS-18	53.90	52.47	143	trc	
			49						less intense fault breccia - Developed in conglomeratic Argillite	MS-19	56.05	53.90	215	0.005	
			50						55.0 - 56.5						
			51												
			52												
			53												
			54												
			55						56.5 - 64.85 SILTY TURBIDITE:	MS-20	58.50	56.05	245	trc	
			56							MS-21	59.00	58.50	670	0.005	
			57						bedding mainly at 85° to c.A., minor short sheared sections throughout.	MS-22	61.70	59.00	250	trc	
			58												
			59												
			60												
			61												
			62												
			63												
			64						64.85 - 67.35 BLEACHED FAULT BRECCIA Green Arg. Blk, Green Wacke, Minor Wacke, ALBITE, ROCK BED/CLV 80% - 90% 89% (17% - 80%) QZ PAR CLV 2.0%; trc (0.90 - 0.5%) AS; 0.19% PR; trc (0.90 - trc) CP; 0.5% PY minor sulfides	MS-23	63.65	61.70	195	0.005	
			65							MS-24	64.85	63.65	120	0.005	
			66												
			67												
			68												
			69												
			70						67.35 - TURBIDITE: greenish, lithic wacke to siltstone Minor Fault 69.8 - 70.35 qtz bx	MS-25	67.65	64.85	280	0.025	
			71												
			72												
			73												
			74												
			75												
			76												
			77												
			78												
			79												
			80												

box 6 not full intentionally

Intense Fault bx

Minor Fault



DIAMOND DRILL RECORD

PROJECT:

HOLE NUMBER: M5 M-5

LOCATION:

INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
										to	from		
71	8						TURBIDITE	67.35 -	M5-27	71.30	70.60	0.60	0.010
72								M5-28	71.85	71.30	0.75	trc	
73								M5-29	72.85	71.95	0.90	0.010	
74													
75													
76	75.6								M5-30	76.30	72.85	3.45	trc
77	9					sheared turbidite 77.3. bx 77.8. gtz bx plus minor pyrite sheared and minor bleaching; alteration		M5-31	77.30	76.30	1.00	trc	
78							M5-32	78.20	77.30	0.90	0.005		
79							M5-33	79.30	78.20	1.10	0.005		
80													



LOCATION (LEVEL):		DIAMOND DRILL RECORD			PROJECT:	HOLE NUMBER: M-6
DIP:	-35°	DIRECTION - 225 TRUE				
LATITUDE:	N	LENGTH:	67.40 meters ()	ELEVATION:	1468.1	CLAIM NUMBER: CARO #3 FRC
DEPARTURE:	E	CORE SIZE:	BQ	DATE LOGGED:		LOCATION:
STARTED:	NOVEMBER 1 1975	FINISHED:	NOVEMBER 2 1975	LOGGED BY:	DJG 1975, JTS 1989	SAMPLED BY:
O.B. THICKNESS:	6.7 meters	STARTED:		FINISHED:		CASING:
B.R. THICKNESS:		STARTED:		FINISHED:		TOTAL RECOVERY: %
CONTRACTOR:		CORE STORED:				

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE COMMENT:	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne		
										to	from				
								0-6.70m OVERBURDEN							
								7.3? start of box							
6.70		7.3						6.70-8.62 CHLORITIC GREYWACKE Quartz-Albite Rock, Interbedded Wackea Argil Bed CLV, 80% CLV/CL/GR 60% INC 80% 10% OZ PHR CLV 20% 3% CA; 0.1% AS; 10% PR; 1% CP; 0.3% PY	M6-1	8.62	6.70	1.92	0.030	} 6.00 m of 0.064 oz/tonne	
								8.62-10.22 ZONE MATERIAL - Dark stained	M6-2	10.22	8.62	1.60	0.01		
								10.22-12.75 Altered + slightly mineralized SILTSTONE -	M6-3	11.12	10.22	0.90	0.040		
								12.75-15.00 Altered siltstone	M6-4	12.70	11.12	1.58	0.075		
		13.5						15.00-15.40 Quartz veining + breccia (Fault zone) graphitic slickensides	M6-5	14.72	12.70	2.02	0.010		
15.20								Interbedded wackea & Argil Bed/CLV/GR 80% 10% (2.30-20%) OZ PHR CLV 20% 3% CA; 0.2% PR; 0.7% PY; 0.9% AS, CP	M6-6	15.20	14.72	0.48	0.010		
								15.40-20.50 CHLORITIC GREYWACKE	M6-7	16.48	15.20	1.28	trc		
									M6-8	17.87	16.48	1.39	trc		
								20.50-26.7 Quartz breccia (Fault) in altered siltstone - turbidite	M6-9	19.60	17.87	1.73	trc		
									M6-10	21.45	19.60	1.85	trc		
								graphitic slickensides at 22.3 to 22.5	M6-11	23.25	21.45	1.80	0.005		
22.4								Graphitic	M6-12	25.45	23.25	2.20	trc		
23.2									M6-13	26.20	25.45	0.75	trc		
									M6-14	27.03	26.20	0.83	trc		
		26.7						Brecciated Fault Breccia in Conglomeratic Argillite	M6-15	28.70	27.03	1.67	trc		
								Brecciated chert bed/CLV/CL 75% 10% (20% 50%) OZ CLV 0%; 1% CA; trc PR; 0.3% PY; 0.2% AS, CP	M6-16	29.50	28.70	0.80	trc		
24.0		4							M6-17	30.20	29.50	0.70	trc		



LOCATION '(LEVEL)'		DIAMOND DRILL RECORD		PROJECT'	HOLE NUMBER'
DIP:	-4	DIRECTION	225° TRUE		M7 M-7
LATITUDE'	4336.7 N	LENGTH'	84.4 m	ELEVATION'	1460.2
DEPARTURE'	2300.2 E	CORE SIZE'		DATE LOGGED'	
STARTED'	November 13 1985	FINISHED'	November 14 1985	LOGGED BY'	
O.B. THICKNESS'		STARTED'		FINISHED'	
B.R. THICKNESS'		STARTED'		FINISHED'	
CONTRACTOR'		CORE STORED'		TOTAL RECOVERY: %	
				SURVEY:	
				DEPTH	BEARING
					ANGLE: Reading Correct

DRILLING INTERVAL	% CORE RECOVERED	BOX Number	SCALE	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE: COMMENT: INTERVAL from to	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne
										to	from		
			1					0 - 25.35 m OVERBURDEN					
			2										
			3										
			4										
			5										
			6										
			7										
			8										
			9										
			10										
			11										
			12										
			13										
			14										
			15										
			16										
			17										
			18										
			19										
			20										
			21										
			22										
			23										
			24										
			25					(0-25.35m) OVB					
25.35		25.2	26					25.35-26.70m PARC SERPENTINITE: dark green	M7-1	26.71	25.35	1.36	0.005
26.71			27					(26.71-40.90m) Quartz Breccia: orange weathering	M7-2	27.54	26.71	0.83	0.055
			28					Quartz-Albite calc. silicate rock, yellow cl. sch. Bed VAR I.C. 60% CLV/GR/SG	M7-3	28.76	27.54	1.22	0.050
			29					5°-60° 13% (32% - 25%) QZ PAR CLV 20% 10% CH 0% AS 0.02% PR etc	M7-4	29.71	28.76	0.95	0.025
			30					CP: 0.4% PY	M7-5	31.45	29.71	1.74	tr



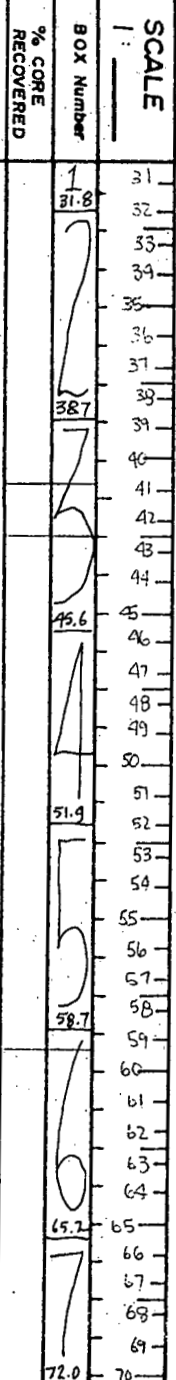
DIAMOND DRILL RECORD

LOCATION :

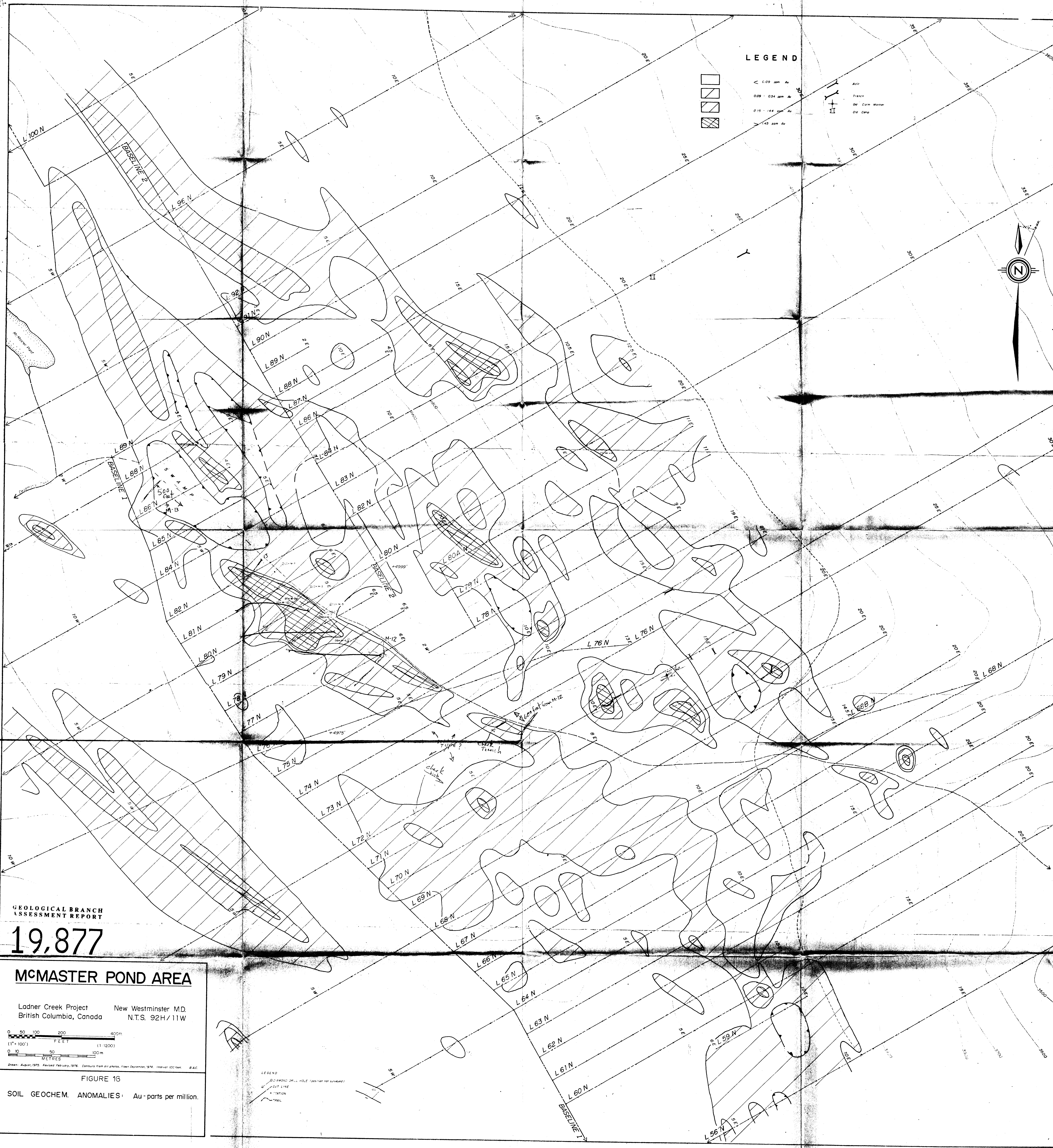
PROJECT :

HOLE NUMBER : M-7

SCALE	ALTERATION	FRACTURING	MINERAL	GEOLOGY	PURPOSE : COMMENT :	SAMPLE NUMBER	METERS		LENGTH METERS	Au g/tonne			
							to	from					
31					26.71 QUARTZ BRECCIA IN CHLORITIC GREYWACKE :	M7-5	31.45	29.71	1.74	trc	13.29 m of 0.045 g/t Au		
32						M7-6	32.45	31.45	1.00	0.020			
33						M7-7	33.00	32.45	0.55	0.075			
34						M7-8	33.85	33.00	0.85	trc			
35						M7-9	34.86	33.85	1.01	0.075			
36						M7-10	36.90	34.86	2.04	0.050			
37						M7-11	38.72	36.90	1.82	0.035			
38					Lots of slickensides, Quartz breccia to 42.54.	M7-12	40.00	38.72	1.28	0.13			
39					(40.90-42.54m) Brecciated Chert, weak CL. SCS; CLV/CL 00°; 20% QZ. PAR. CLV; 0% CA; 0.1% AS; trc PR; 0.4% PY; 0% CP	M7-13	40.90	40.00	0.90	0.020			
40						M7-14	41.40	40.90	0.50	trc			
41						M7-15	42.54	41.40	1.14	0.010			
42					(42.54 to 59.40m) Interbedded Argill. Wacke; Bed. CLV/CL/G-R INC 80°; 12% QZ. PAR. CLV; trc CA; 0.2% PR; 0.7% PY; AS, CP	M7-16	44.20	42.54	1.66	0.005			
43					altered siltstone	M7-17	45.40	44.20	1.20	0.005			
44						M7-18	46.25	45.40	0.85	0.005			
45						M7-19	47.17	46.25	0.92	0.005			
46						M7-20	49.10	47.17	1.93	trc			
47					Quartz breccia 49.0 - 58.5, brown-orange weathering	M7-21	50.95	49.10	1.75	0.020			
48						M7-22	51.90	50.95	1.05	0.010			
49						M7-23	53.90	51.90	2.00	0.010			
50						M7-24	54.71	53.90	0.81	trc			
51					Graphitic slickensides 57.2 - 58.5	M7-25	57.90	54.71	3.19	0.005			
52					58.5- TURBIDITE	M7-26	59.40	57.90	1.50	0.005			
53					(59.40-84.90m) Green Argill. Minor Quartz - Albite. Bed. CLV 75° INC 85°; 3% QZ. PAR. CLV. 0.0% trc. CA; trc. CO 2% - trc. AS; trc. PR; 0.2% PY; 0% CP	M7-27	60.75	59.40	1.35	0.040			
54						M7-28	62.77	60.75	2.02	trc			
55						M7-29	63.47	62.77	0.70	trc			
56						M7-30	65.95	63.47	2.48	trc			
57						M7-31	68.50	65.95	2.55	trc			
58						M7-32	70.70	68.50	2.20	trc			



40.90
47.59
54.40

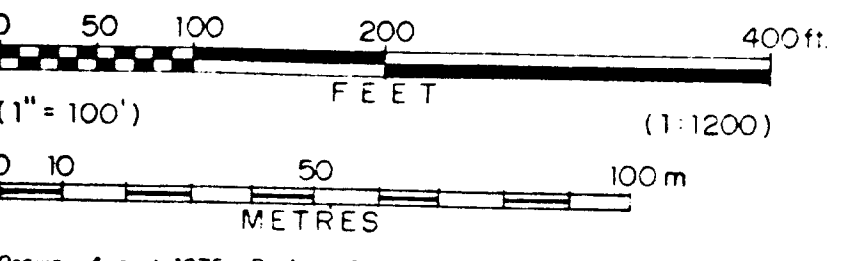


GEOLOGICAL BRANCH
ASSESSMENT REPORT

19,877

MCMMASTER POND AREA

Ladner Creek Project New Westminster MD
British Columbia, Canada N.T.S. 92H/11W



Drawn: August, 1975. Revised: February, 1976. Contours from air photos, film September, 1974. Interval 100 feet. B.A.C.

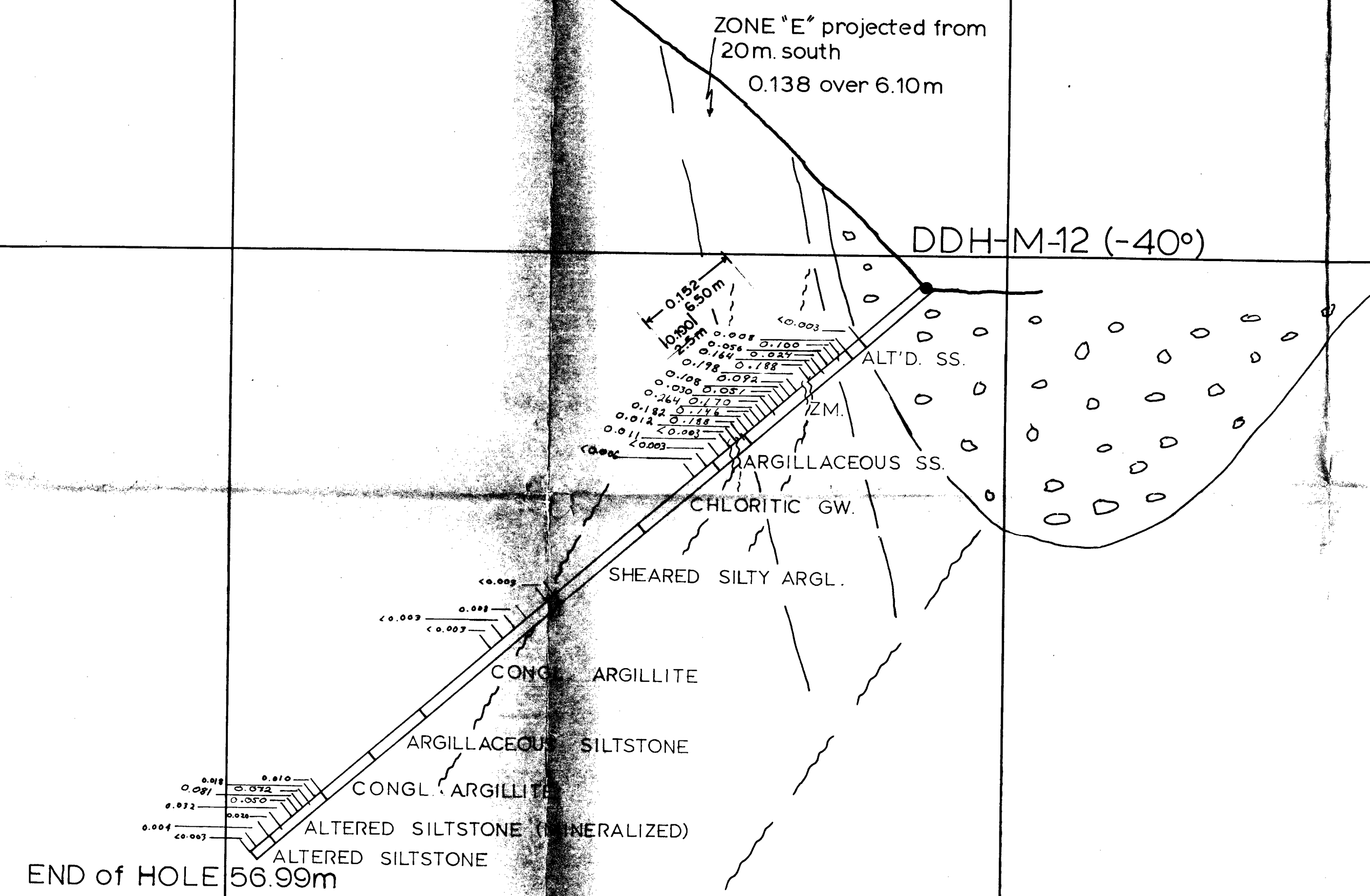
FIGURE 16
SOIL GEOCHEM. ANOMALIES: Au - parts per million.

LEGEND
 (Symbol) BOREHOLE DRILL HOLE (position not surveyed)
 (Symbol) CUT LINE
 (Symbol) STATION
 (Symbol) TRAIL

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430
1420



END of HOLE 56.99m

DDH-M-12 (-40°)

ZONE "E" projected from 20m. south 0.138 over 6.10m

1+50W 1+40W 1+30W 1+20W 1+10W 1+00W 0+90W 0+80W 0+70W 0+60W 0+50W 0+40W 0+30W 0+20W 0+10W 0+00 0+10E 0+20E 0+30E 0+40E 0+50E 0+60E

STATIONS (10 m intervals)

ABBREVIATIONS
 CHL. CHLORITE
 SS. SILTSTONE
 ALT'D. ALTERED
 CONGL. CONGLOMERATIC
 Z.M. ZONE MATERIAL
 ARGL. ARGILLITE
 GW. GYPSUM
 W. LITHIC WACKE
 BLDR. BOULDER
 QTZ. QUARTZ
 MIN'R'L'D. MINERALIZED

GEOLOGICAL BRANCH ASSESSMENT REPORT

19,877

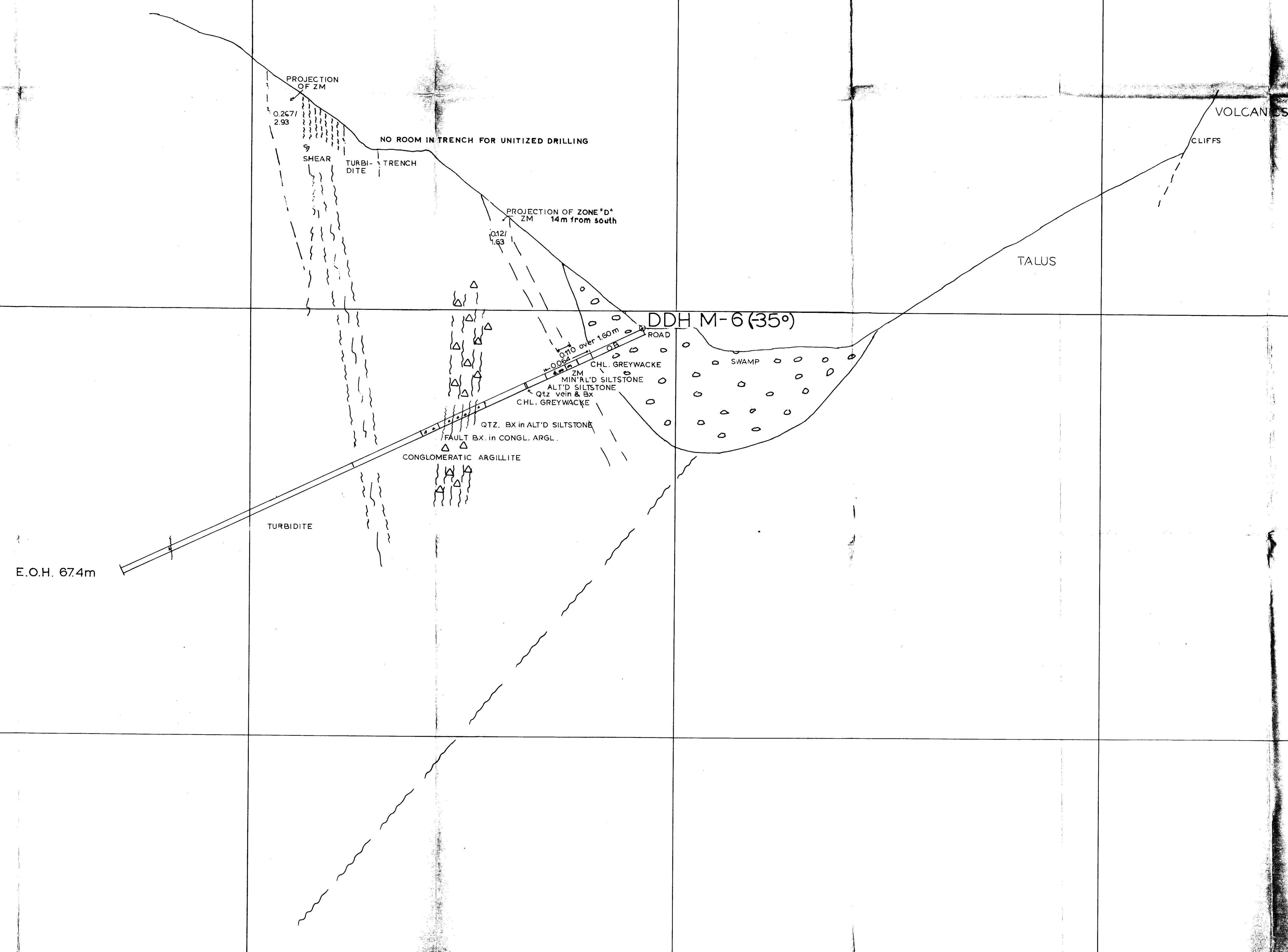
0 10 20m

CAROLIN MINES LTD.
 SCALE: 1:250 ENG. JTS. WBL
 DATE: November, 1989 FIGURE 15
 McMASTER ZONE GEOLOGY CROSS-SECTION
 DRILLHOLE M-12
 NEW GLOBAL RESOURCES LTD.

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430



1+50W 1+40W 1+30W 1+20W 1+10W 1+00W 0+90W 0+80W 0+70W 0+60W 0+50W 0+40W 0+30W 0+20W 0+10W 0+00 0+10E 0+20E 0+30E 0+40E 0+50E 0+60E

STATIONS (10m intervals)

ABBREVIATIONS
 CHL. CHLORITE
 SS SILTSTONE
 ALT'D ALTERED
 CONGL. CONGLOMERATIC
 ZM ZONE MATERIAL
 ARGL. ARGILLITE
 GW GREYWACKE
 L.W. LITHIC WACKE
 BLDR BOULDER
 QZ QUARTZ
 MIN RLD MINERALIZED

GEOLOGICAL BRANCH
ASSESSMENT REPORT

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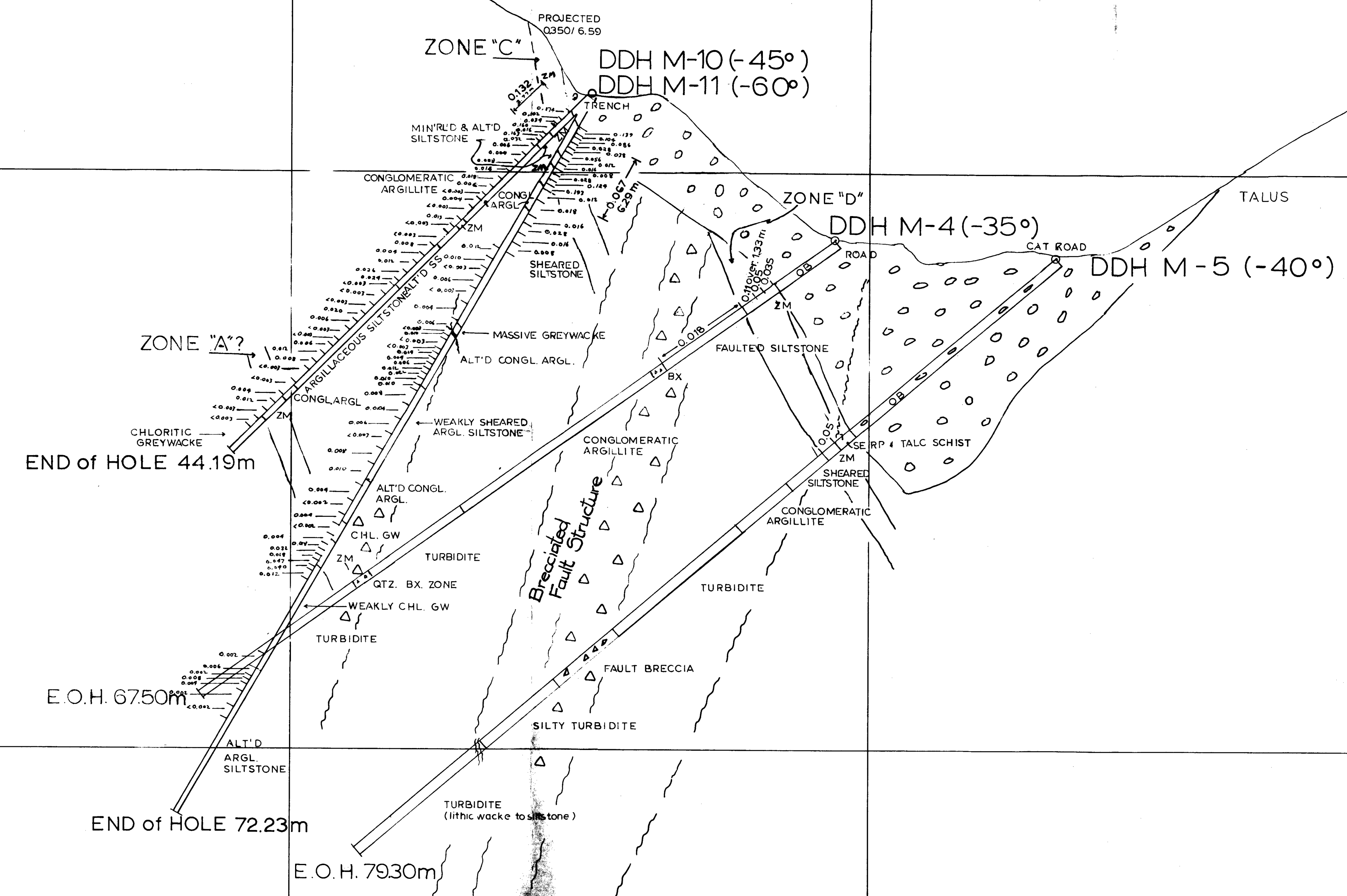
0 10 20m

CAROLIN MINES LTD.
 SCALE: 1:250 ENG. JTS., WBL
 DATE: October 1989 FIGURE 14
 McMASTER ZONE GEOLOGY CROSS-SECTION
 DRILL HOLE M-6
 NEW GLOBAL RESOURCES LTD.

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430



1+50W 1+40W 1+30W 1+20W 1+10W 1+00W 0+90W 0+80W 0+70W 0+60W 0+50W 0+40W 0+30W 0+20W 0+10W 0+00 0+10E 0+20E 0+30E 0+40E 0+50E 0+60E

STATIONS (10m intervals)

GEOLOGICAL BRANCH
ASSESSMENT REPORT

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0 10 20m

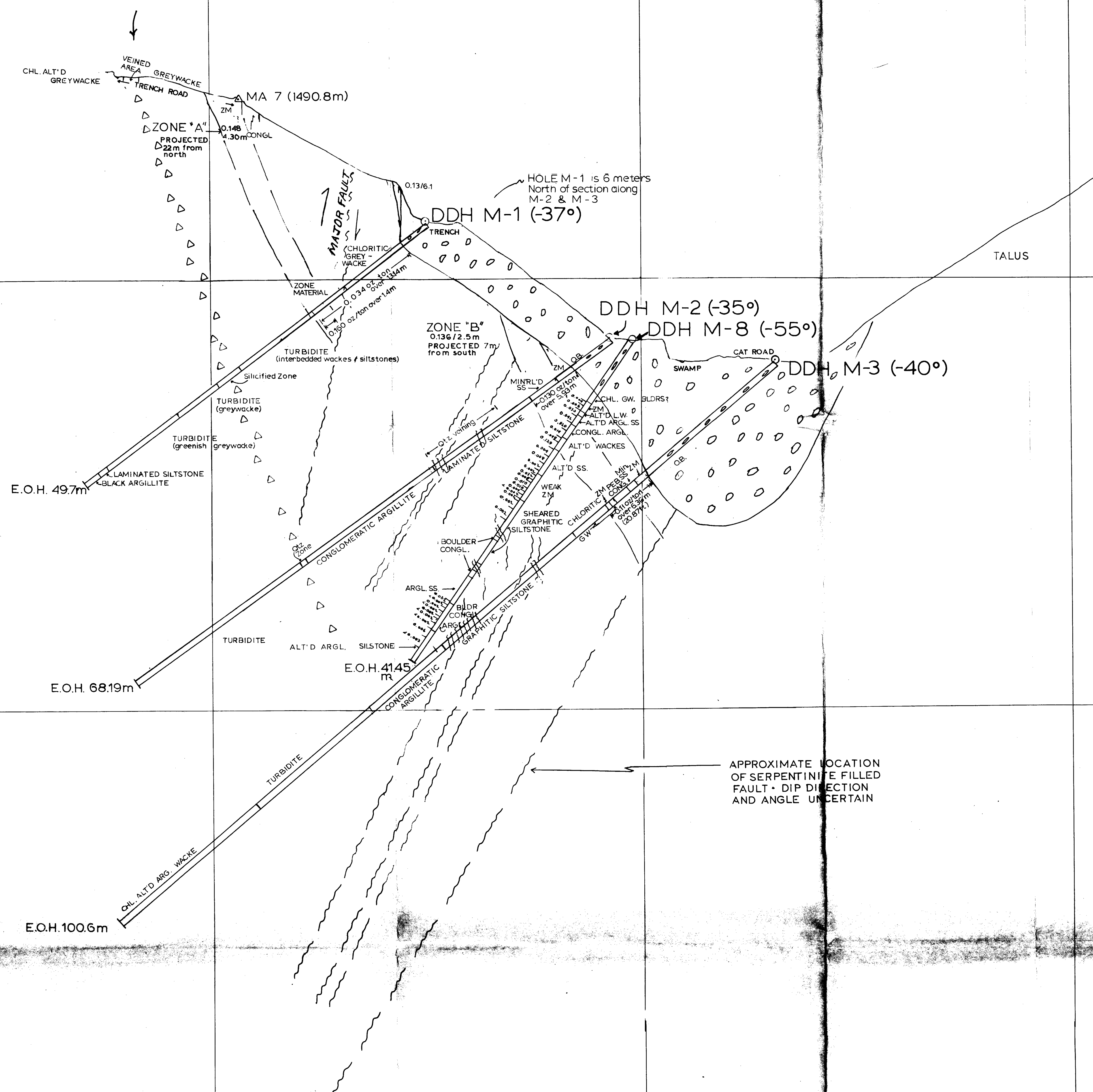
ABBREVIATIONS
 CHL. CHLORITE
 SS SILTSTONE
 ALT'D ALTERED
 CONGL. CONGLOMERATIC
 ZM ZONE MATERIAL
 ARGL. ARGILLITE
 GW GREYWACKE
 LW LITHIC WACKE
 BLDR BOULDER
 QTZ QUARTZ
 MIN'RL'D MINERALIZED

CAROLIN MINES LTD.
 SCALE: 1:250 ENG. JTS., WBL
 DATE: October, 1989 FIGURE 13
 McMASTER ZONE GEOLOGY CROSS-SECTION
 DRILL HOLES M-4 & M-5, M-10 & M-11
 NEW GLOBAL RESOURCES LTD.

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430



GEOLOGICAL BRANCH
ASSESSMENT REPORT

19,877

0 10 20m

ABBREVIATIONS
 CHL. CHLORITE
 SS. SILTSTONE
 ALT'D. ALTERED
 CONGL. CONGLOMERATIC
 ZM. ZONE MATERIAL
 ARGL. ARGILLITE
 GW. GREYWACKE
 LW. LITHIC WACKE
 BLD. BOULDER
 QZ. QUARTZ
 MINRLD. MINERALIZED

CAROLIN MINES LTD.
 SCALE: 1:250 | ENG: JTS, WBL
 DATE: October 1989 | FIGURE 12
 McMASTER ZONE - GEOLOGY CROSS-SECTION
 DRILL HOLES M-1, M-2 & M-3, M-8
 NEW GLOBAL RESOURCES LTD.

STATION (10m intervals)

1+50W 1+40W 1+30W 1+20W 1+10W 1+00W 0+90W 0+80W 0+70W 0+60W 0+50W 0+40W 0+30W 0+20W 0+10W 0+00 0+10E 0+20E 0+30E 0+40E 0+50E 0+60E

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430



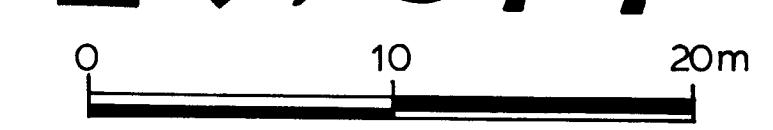
1+59W
1+40W
1+30W
1+20W
1+10W
1+00W
0+90W
0+80W
0+70W
0+60W
0+50W
0+40W
0+30W
0+20W
0+10W
0+00
0+10E
0+20E
0+30E
0+40E
0+50E
0+60E

STATIONS (10m intervals)

ABBREVIATIONS
OB OVERBURDEN
ALT'D ALTERED

GEOLOGICAL BRANCH
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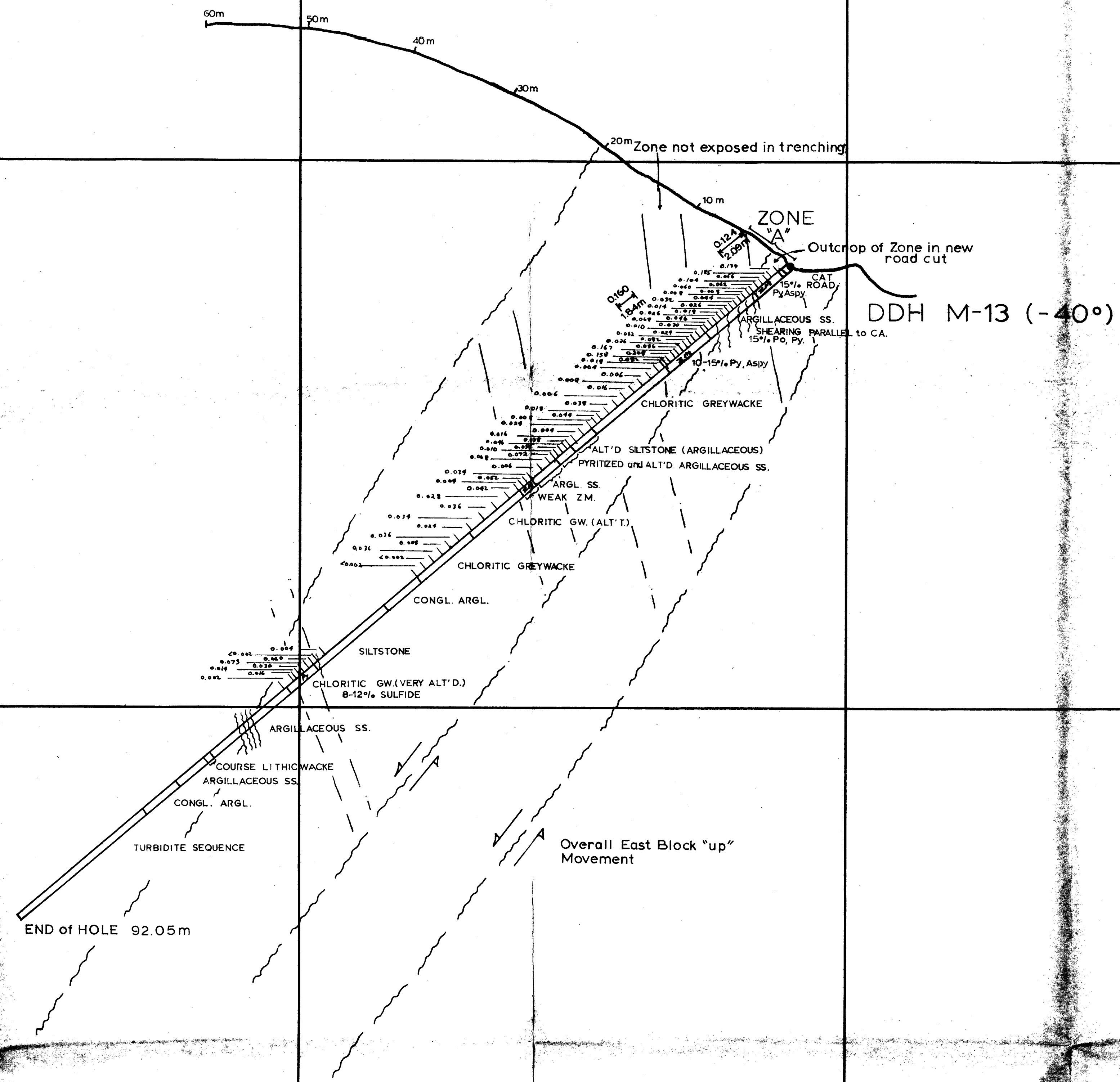


CAROLIN MINES LTD.	
SCALE: 1:250	ENG: JTS, WBL
DATE: October 1989	FIGURE 11
McMASTER ZONE- GEOLOGY CROSS-SECTION DRILL HOLE M-7, M-9	
NEW GLOBAL RESOURCES LTD.	

VIEW LOOKING TOWARDS AZIMUTH 315°

ELEVATION (m)

1510
1500
1490
1480
1470
1460
1450
1440
1430
1420



END OF HOLE 92.05m

Overall East Block "up" Movement

DDH M-13 (-40°)

ZONE A

Outcrop of Zone in new road cut

20m Zone not exposed in trenching

STATIONS (10m intervals)

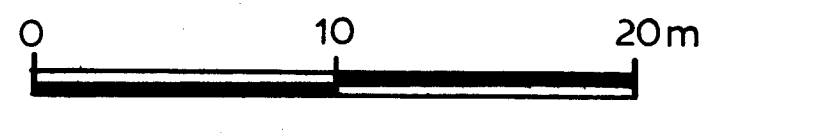
1+50W
1+40W
1+30W
1+20W
1+10W
1+00W
0+90W
0+80W
0+70W
0+60W
0+50W
0+40W
0+30W
0+20W
0+10W
0+00
0+10E
0+20E
0+30E
0+40E
0+50E
0+60E

ABBREVIATIONS

CHL	CHLORITE
SS	SILTSTONE
ALT'D	ALTERED
CONGL.	CONGLOMERATIC
Z.M.	ZONE MATERIAL
ARGL.	ARGILLITE
GW.	GREYWACKE
LW.	LITHIC WACKE
Bldr.	BOULDER
QTZ	QUARTZ
MIN'R'L'D.	MINERALIZED

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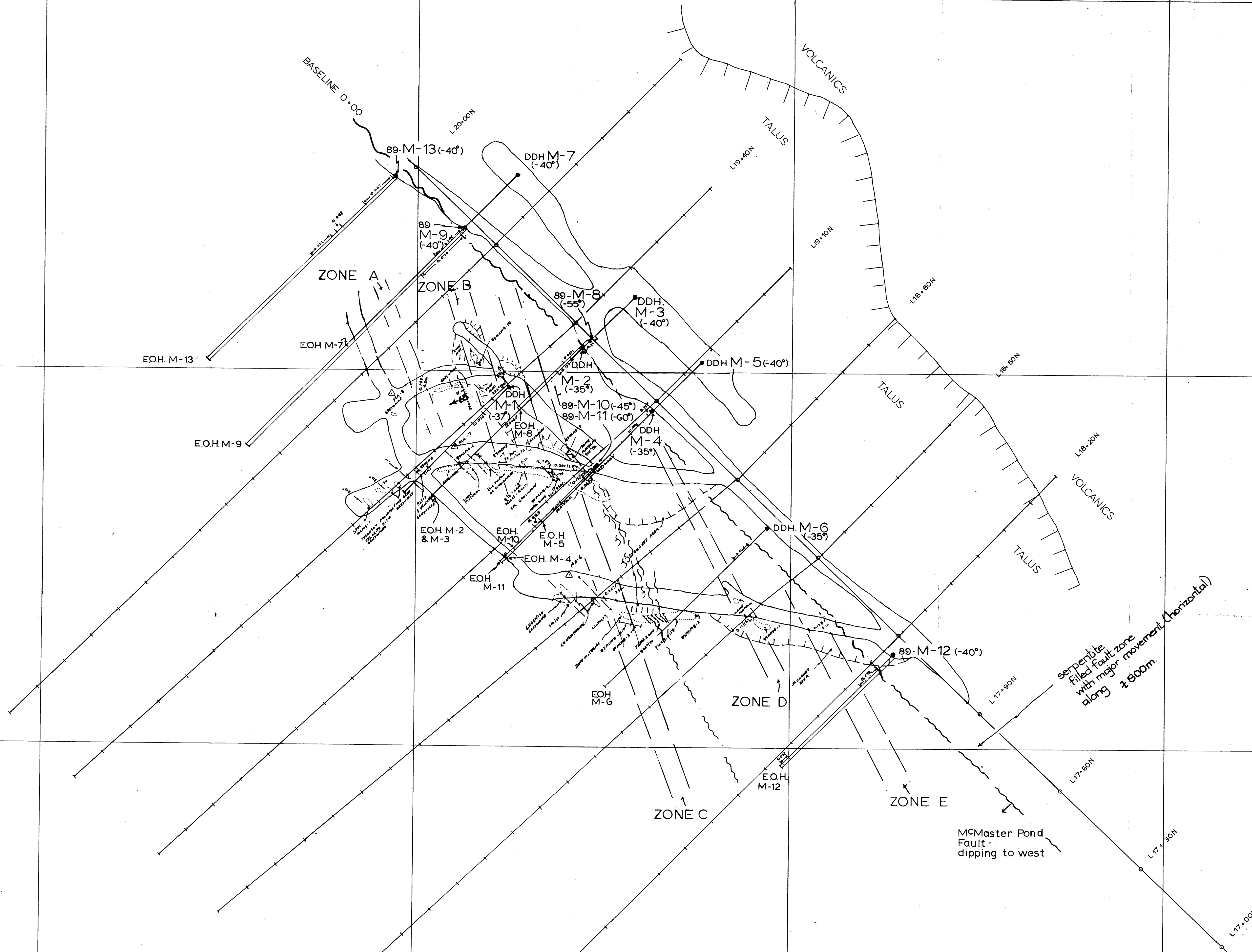
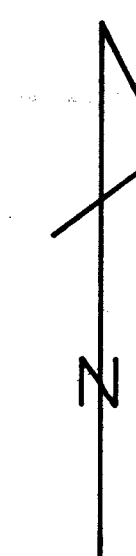
CAROLIN MINES LTD.

SCALE: 1:250 ENG. JTS, WBL

DATE: November, 1989 FIGURE 10

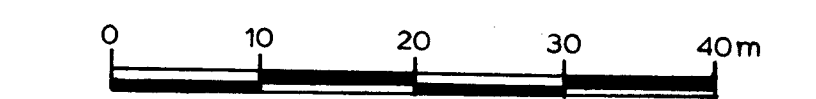
McMASTER ZONE-GEOLOGY CROSS-SECTION
DRILLHOLE M-13

NEW GLOBAL RESOURCES LTD.

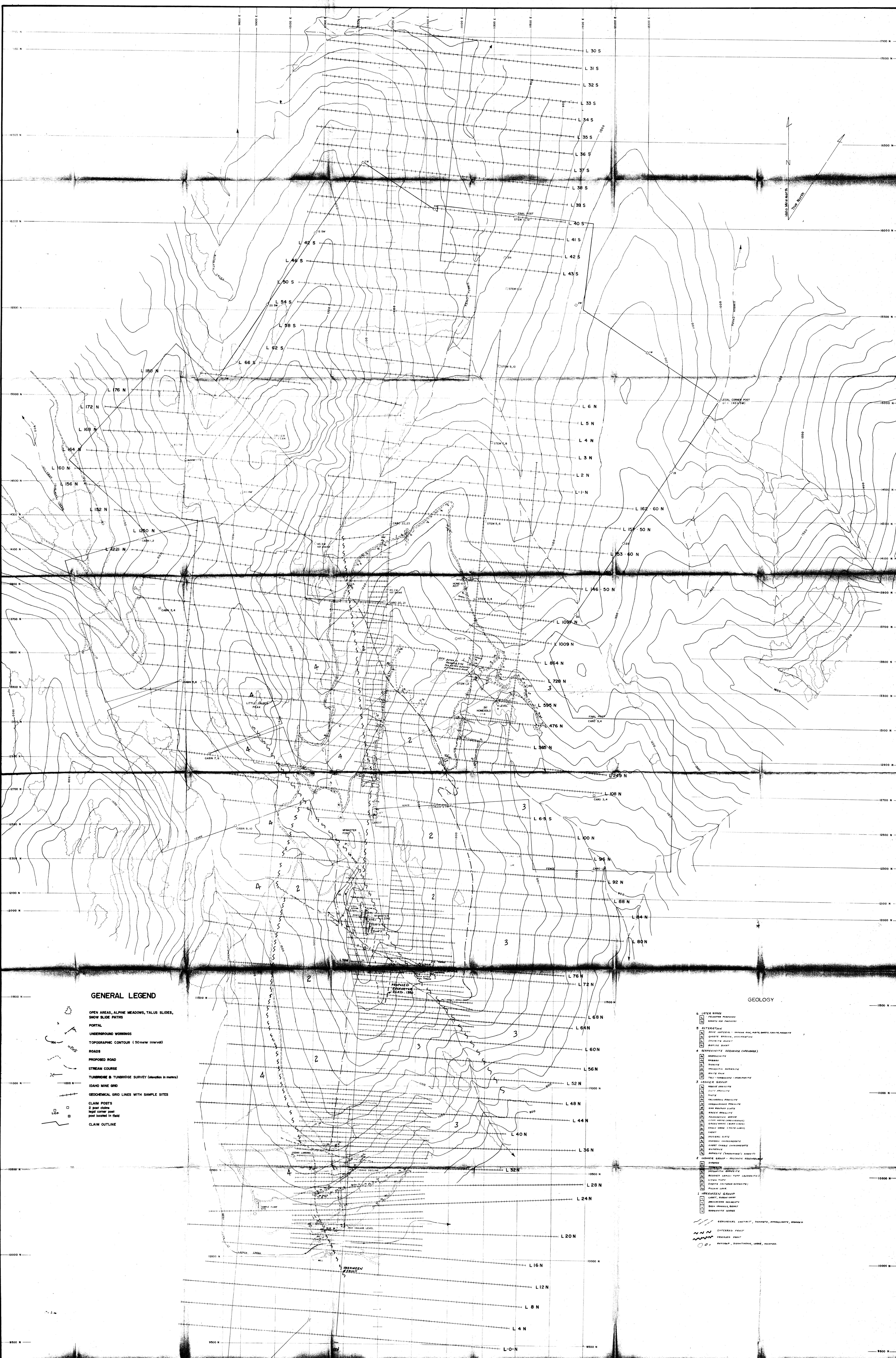


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CAROLIN MINES LTD.	
SCALE: 1: 500	ENG. WBL, JTS
DATE: October 1989	FIGURE 9
McMASTER ZONE - GEOLOGY PLAN	
NEW GLOBAL RESOURCES LTD.	



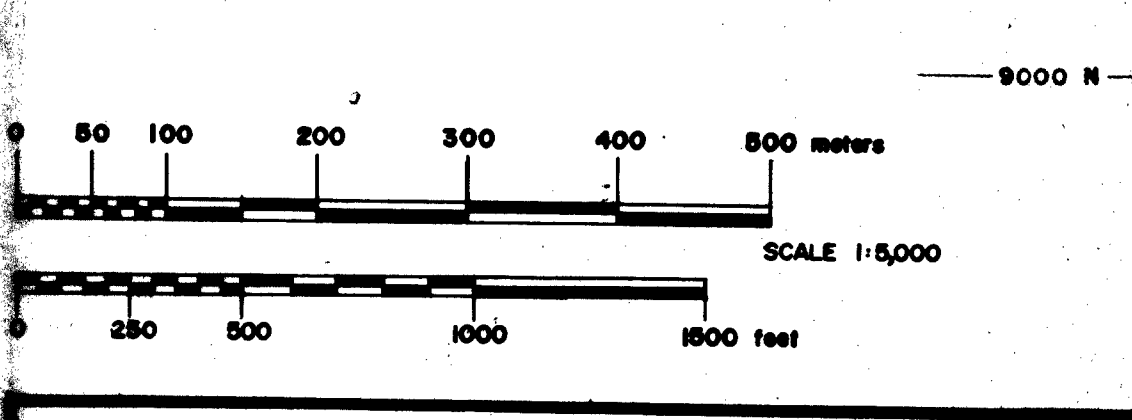
GENERAL LEGEND

- OPEN AREAS, ALPINE MEADOWS, TALUS SLIDES, SNOW SLIDE PATHS
- PORTAL
- UNDERGROUND WORKINGS
- TOPOGRAPHIC CONTOUR (50m interval)
- ROADS
- PROPOSED ROAD
- STREAM COURSE
- TUNNAGE & TUNNAGE SURVEY (elevation in meters)
- IDAHO MINE GRID
- GEOCHEMICAL GRID LINES WITH SAMPLE SITES
- CLAIM POSTS
- HIGH CORNER POST
- POST NUMBER IN FIELD
- CLAIM OUTLINE

GEOLOGY

- 6 LATER BYRDS
- 5 ALTERATION
- 4 BERRINGTON'S SEQUENCE (CAMPBELL)
- 3 LADNER GROUP
- 2 LADNER GROUP - VOLCANIC SEQUENCE
- 1 UNCONFORMABLE GROUP
- 0.5 UNCONFORMABLE GROUP
- GEOLOGICAL CONTACT, UNCONFORMITY, DISCONFORMITY
- EXTENDED FAULT
- FAULTED FAULT
- ○ ○ OUTCROP, QUANTIFICATION, LENS, MOUNTAIN

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CAROLIN MINES LIMITED
LADNER CREEK NORTH PROJECT

**GEOLOGY
OF THE CENTRAL COQUHALLA GOLD BELT
COMPOSITE MAP**

SCALE 1:5000
DATE 02H 11/W
DRAWN BY RKB, JFS, DM
CHECKED BY JFS
DATE FEB 2/82
FIGURE 7
FIGURE 7