

A SUMMARY REPORT ON THE
J & L MINERAL OPTION
LEAD-ZINC-GOLD-SILVER PROSPECT
BRITISH COLUMBIA
NTS: 82M/8E

For: Selco Division - BP Resources Canada Limited
Vancouver, BC

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BPVR 84-53

February 27, 1985

RESUME

The J & L property, which is located some 32 km north of Revelstoke, BC, contains 6 or more arsenical lead-zinc-silver-gold zones within metasediments of the Hamill Group (Lower Cambrian).

The summary of the 1984 results is as follows:

1. The J & L occurrence still appears to be a sedex-type, arsenical deposit.
2. The J & L Main Zone massive sulphide sheet, which is hosted by isoclinally folded metasediments and sediments, has undergone a long and complex structural history with at least 5 phases of deformation. It is believed that this is a structural overprint on the sulphide system which has been controlled by the stratigraphy.
3. Surface mapping and sampling has extended the known arsenical mineralization for 1.54 km to the northwest of the 830 m level portals and is still "open" to the north. This mineralization appears to be the strike extension of the J & L Main Zone and increases the entire, observed strike length to over 3.34 km.

4. A total of 40 new arsenopyrite-pyrite (+ sphalerite and occurrences galena) were located in 4 parallel sub-zones within the sericitic quartzites of the Hamill Group. The surface expression of the North Zone mineralization is relatively narrow and discontinuous, with the most significant result being 0.27% Pb, 0.09% Zn, 1.881% As, 20.8 g/t Ag and 3.1 g/t Au over an estimated true width of 2.11 metres.
5. The new drifting on the 830 m level has followed the Main Zone for, approximately, 316 m beyond the 1983 work. The development has now followed the Main Zone for 810 m, which is, at most, 43% of the down dip extension of the surface expression of the Main Zone on Goat Mountain.
6. The 1984 drifting indicated 2 separate mineralized zones with a narrower, lower grade intervening section. The weighted average grades of the drift face sampling and the short drill hole program are as follows:
 - a) Zone 1: 1.61% Pb, 3.12% Zn, 4.999% As, 48.1 g/t Ag and 7.4 g/t Au over an estimated true width of 2.51 m and length of 161.663 m.

- b) Intervening Section: 0.46% Pb, 0.80% Zn, 5.284% As, 17.0 g/t Ag and 3.4 g/t Au over an estimated true width of 1.30 m and length of 28.188 m.
- c) Zone 2: 2.89% Pb, 5.71% Zn, 2.306% As, 55.4 g/t Ag and 3.1 g/t Au over an estimated true width of 2.37 m and length of 97.005 m (open along strike).
- d) Overall 1984 Drifting: 1.86% Pb, 3.71% Zn, 3.839% As, 45.6 g/t Ag and 5.4 g/t Au over an estimated true width of 2.46 m and length of 291.528 m.

Since the drift did not always intersect the entire lateral extent of the Main Zone, the above results maybe, in part, misleading.

7. The tenor of the mineralization accounted for by only the short drill hole program and their corresponding drift faces is as follows:

- a) Zone 1: 1.67% Pb, 3.11% Zn, 5.028% As, 43.6 g/t Ag and 7.3 g/t Au over an estimated true width of 4.09 m and length of 143.404 m.

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INTRODUCTION

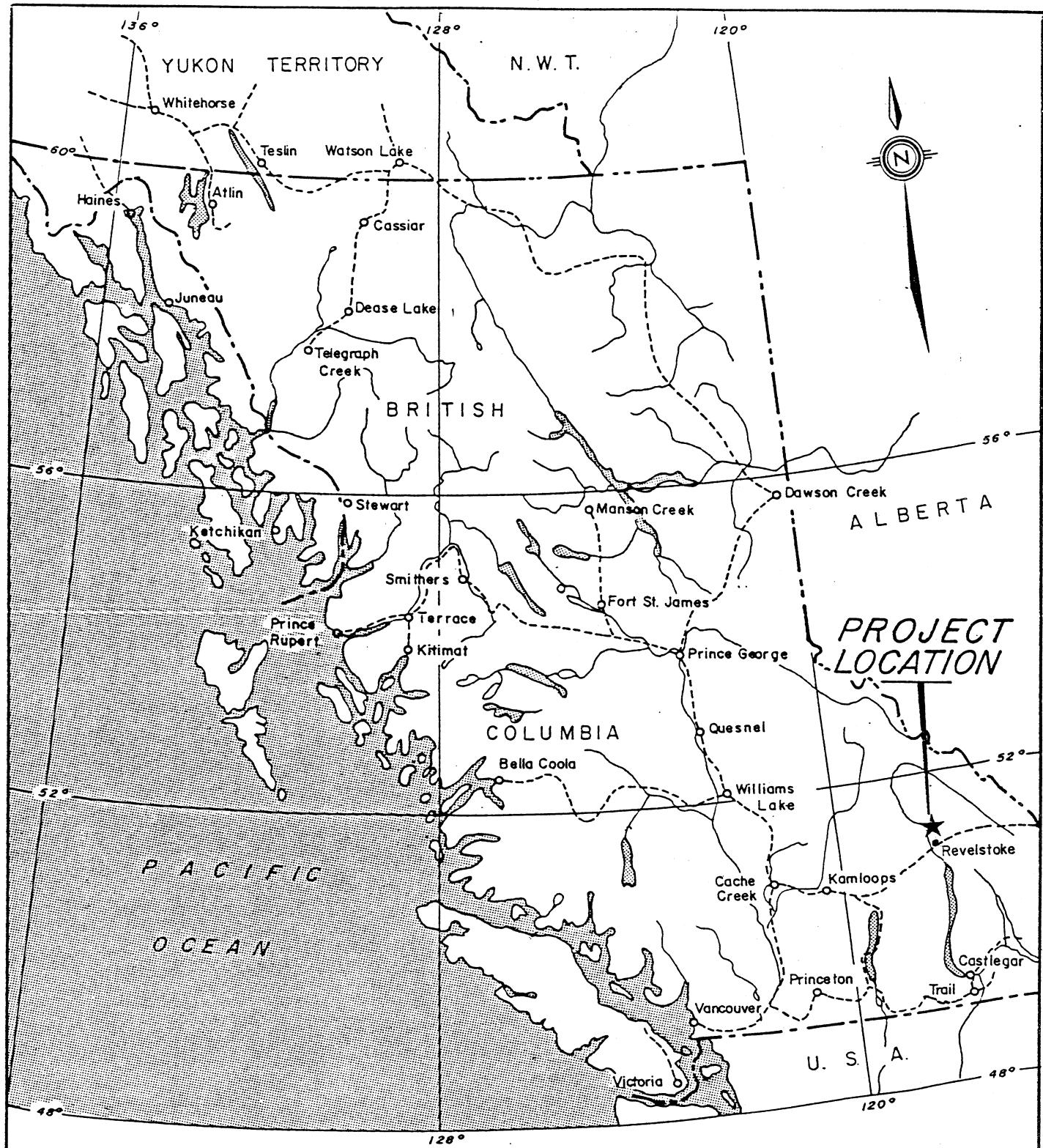
During 1984, a Selco field crew completed a geology and geochemistry surface survey of part of the project area and a detailed geological and geochemical underground study of part of the J & L Main Zone. This study including drifting along the zone, mapping and sampling of the zone and subsequent underground drilling.

The exploration target in this area is economic arsenical lead-zinc-gold-silver, massive sulphide, mineralization.

1. Location, Access, Physiography and Climate

The property is located along and north of Carnes Creek, approximately 32 air km north of the town of Revelstoke (See Figures 1 and 2), at latitude 51°17'N and longitude 118°08'W.

Access is provided by approximately 35 km of paved road (Highway #23), and then a rough 10 km bush road to the property. Helicopter service is also available from Revelstoke. A rough 4-wheel drive road and several overgrown walking trails are found within the property.



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J & L PROJECT
LOCATION MAP

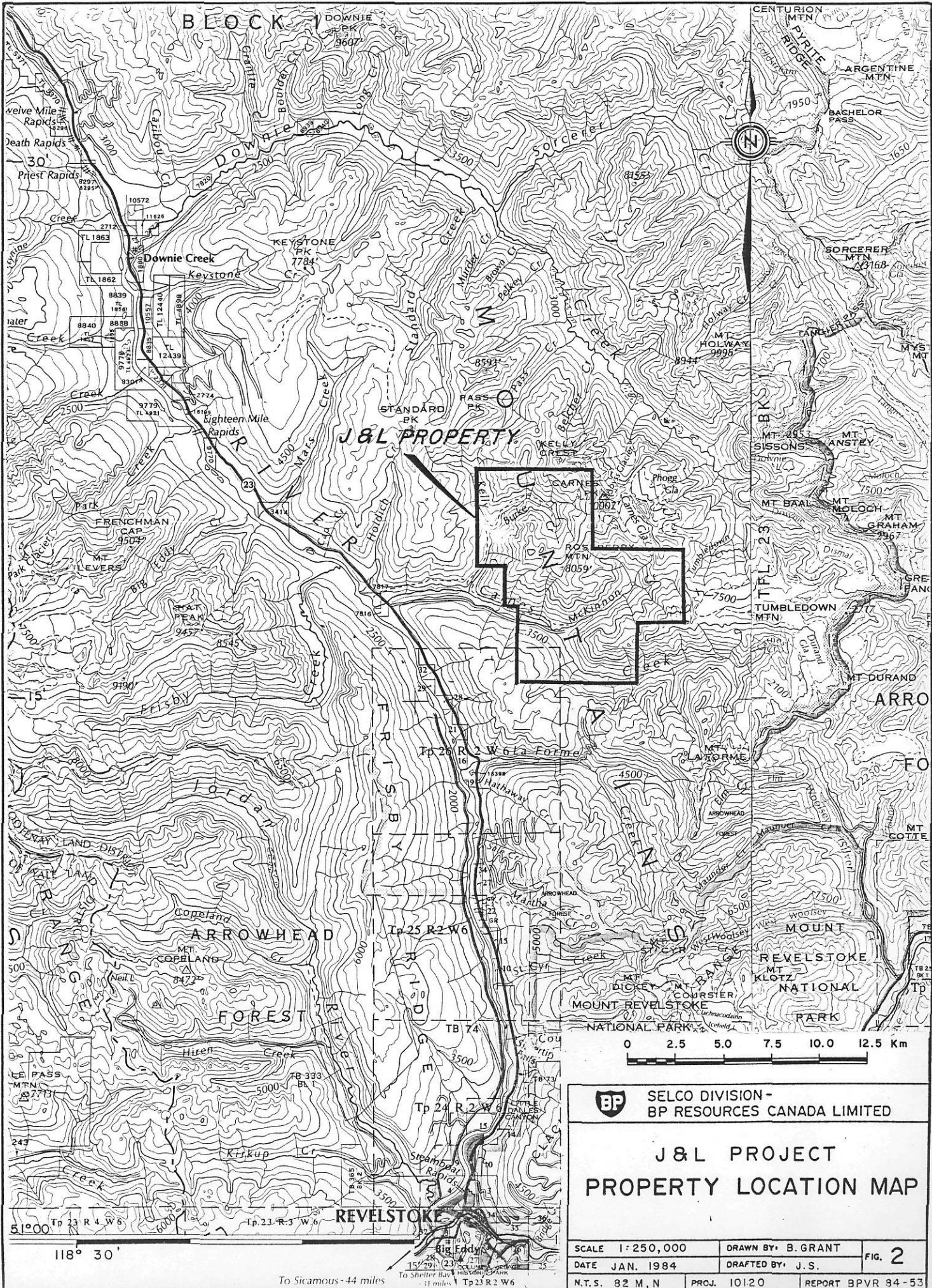
SCALE 1:1,000,000	DRAWN BY R. PEGG	FIG. 1
DATE NOV. 1984	DRAFTED BY J. S.	
V.T.S.	PROJ. 10123	REPORT BPVR 84-53

Maximum relief in the area of the property is 2349 m (3050 to 701 m). The J & L adits are found at the 830 metre elevation and the 986 metre elevation respectively, and are accessible by road and/or trail. Access throughout most of the property is difficult and slow as the steep-sided valleys generally obtain slopes 30 to 40 degrees and are densely covered with rotting cedar and hemlock trees. Locally, windfall, deadfall, alders, devils club, stinging nettles and second growth are extensive. Treeline is at, approximately, the 1980 metre elevation and permanent glaciers are found above 2286 metres. The property is fairly well drained on the south and east by Carnes and McKinnon Creeks and on the west by Kelly Creek.

Climatic conditions dictate a June to October field season. The winters are long and relatively mild with snowfall of between 1 and 4 metres. The mountainous terrain results in numerous snow and earth slides. The summers "usually" have a medium rainfall and temperatures range from 16 to 30 degrees centigrade.

2. Property Status

The property consists of 10 crown granted claims and 32 claim blocks (357 mineral claim units, See Figure 3). Pivak Explorco Limited of Toronto is the owner.



The mineral claims and blocks have been placed into 5 mineral claim groups and consist of the following:

Arty #1 Group (90 Units)

<u>Name</u>	<u>Record No.</u>	<u>Date Recorded</u>	<u>No. Units</u>	<u>Expiry Date</u>
L14827 (View Fraction)				
L14829 (Creek Fraction)				
Arty 1	1219	June 10, 1981	12	June 10, 1989*
Arty 2	1220	June 10, 1981	20	June 10, 1989*
G.D.	603	Apr. 17, 1979	16	Apr. 17, 1993*
Min	604	Apr. 17, 1979	8	Apr. 17, 1988*
Tom	605	Apr. 17, 1979	20	Apr. 17, 1994*
Mary 5	758	Oct. 10, 1979	1	Oct. 10, 1988*
Mary 6	759	Oct. 10, 1979	1	Oct. 10, 1988*
Mary 7	760	Oct. 10, 1979	1	Oct. 10, 1988*
Burke 1	1485	Sep. 30, 1982	9	Sep. 30, 1987*

Arty #3 Group (87 Units)

<u>Name</u>	<u>Record No.</u>	<u>Date Recorded</u>	<u>No. Units</u>	<u>Expiry Date</u>
L14821 (Goat Fraction)				
L14822 (Goat No.2 Fraction)				
L14823 (Goat No.3 Fraction)				
L14824 (Goat No.4 Fraction)				
L14825 (Goat No.5 Fraction)				
L14826 (Goat No.6 Fraction)				
L14828 (View No.2 Fraction)				
Shannon 400	1143	Dec. 17, 1980	20	Dec. 17, 1993*
Shannon 500	1144	Dec. 17, 1980	20	Dec. 17, 1993*
Arty 3	1221	June 10, 1981	20	June 10, 1989*
Arty 4	1222	June 10, 1981	20	June 10, 1989*

Tom Group (28 Units)

<u>Name</u>	<u>Record No.</u>	<u>Date Recorded</u>	<u>No. Units</u>	<u>Expiry Date</u>
Shannon 100	1140	Dec. 17, 1980	12	Dec. 17, 1993*
Shannon 300	1142	Dec. 17, 1980	16	Dec. 17, 1993*

Sam Group (83 Units)

<u>Name</u>	<u>Record No.</u>	<u>Date Recorded</u>	<u>No. Units</u>	<u>Expiry Date</u>
Sam	1549	Nov. 30, 1982	8	Nov. 30, 1993*
Sam 1	1550	Nov. 30, 1982	8	Nov. 30, 1993*
Mary	1545	Nov. 30, 1982	1	Nov. 30, 1993*
Mary 1	1546	Nov. 30, 1982	1	Nov. 30, 1993*
Mary 2	1547	Nov. 30, 1982	1	Nov. 30, 1993*
Mary 3	1548	Nov. 30, 1982	1	Nov. 30, 1993*
Mary 4	757	Oct. 10, 1982	1	Oct. 10, 1988*
Shannon 700	1146	Dec. 17, 1980	18	Dec. 17, 1988*
Shannon 800	1147	Dec. 17, 1980	8	Dec. 17, 1988*
Shannon 900	1148	Dec. 17, 1980	20	Dec. 17, 1988*
Shannon 1000	1149	Dec. 17, 1980	10	Dec. 17, 1988*
Shannon 1100	1150	Dec. 17, 1980	6	Dec. 17, 1988*

Burke Group (79 Units)

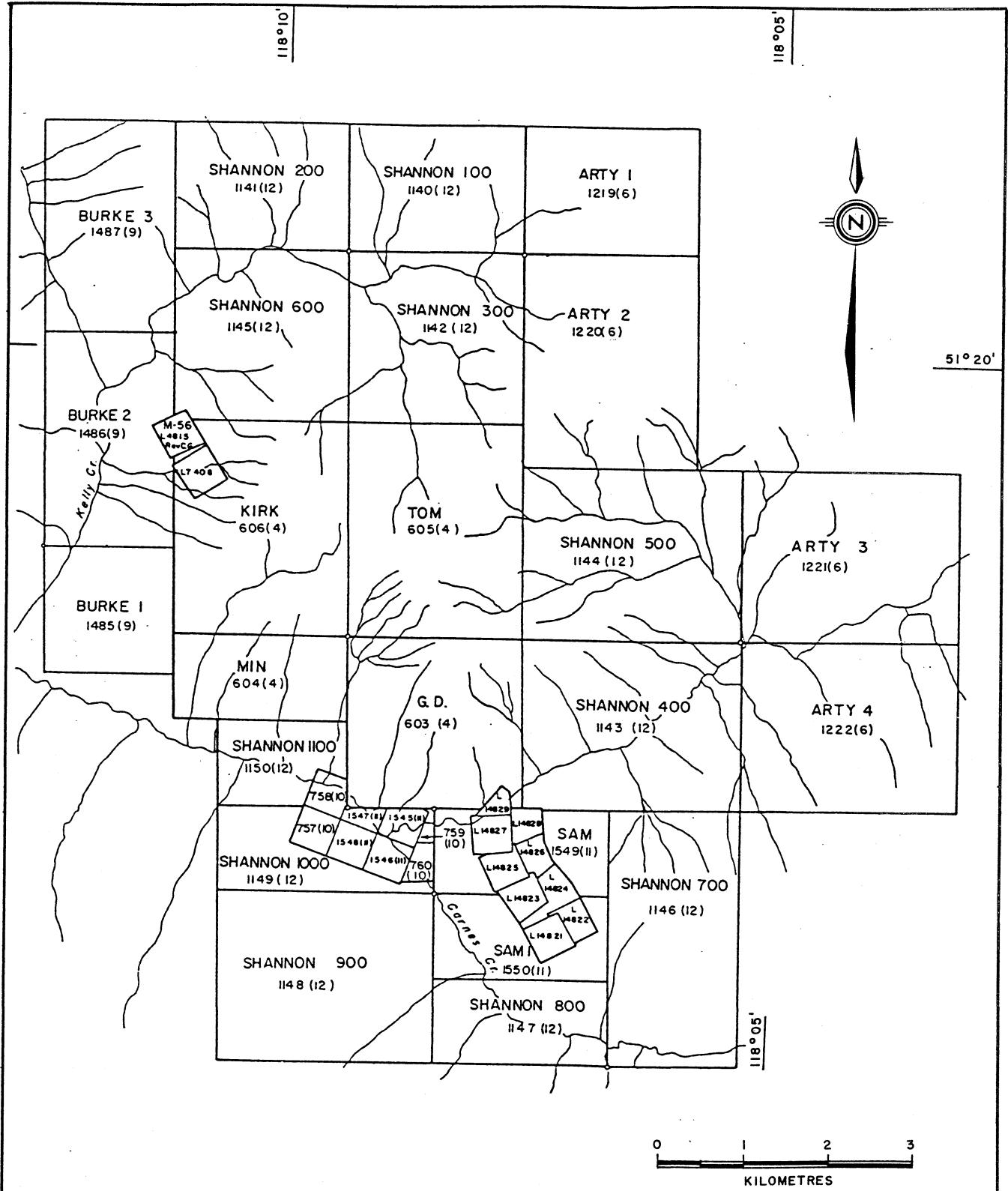
<u>Name</u>	<u>Record No.</u>	<u>Date Recorded</u>	<u>No. Units</u>	<u>Expiry Date</u>
L7408 (Aberdeen)				
Kirk	606	Apr. 17, 1979	20	Apr. 17, 1993*
Shannon 200	1141	Dec. 17, 1980	12	Dec. 17, 1993*
Shannon 600	1145	Dec. 17, 1980	16	Dec. 17, 1993*
Burke 2	1486	Sep. 30, 1982	15	Sep. 30, 1986*
Burke 3	1487	Sep. 30, 1982	15	Sep. 30, 1986*

*Assessment approval pending

PLEASE NOTE: L4815 (Hardpan) is a reverted crown grant which is now a Mineral Lease (M.R. #56) and is owned by a James H. Elliot who resides at 910 - 3rd Street West, Revelstoke, BC.

3. History of Exploration

The area first became an exploration target after 1865 when placer gold was discovered in Carnes Creek and other creeks



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J & L PROJECT
CLAIM MAP

SCALE As Shown	DRAWN BY: R.PEGG	FIG. 3
DATE NOV. 1984	DRAFTED BY: J.S.	
N.T.S. 82 M/8E	PROJ. 10120	REPORT BPVR 84-53

in the vicinity. Prospecting began in the late 1800's and the J & L Prospect was first staked in 1896. The area then was intensely prospected and by the early 1900's, there were over 20 base and precious metal prospects discovered. Three of these, the J & L (Au-Ag-Pb-Zn), the Roseberry (Au) and the A & E (Zn-Pb-Ag), are found within the property boundary.

The property was worked, intermittently, from the time of discovery to 1942. During this time, the underground development on the J & L consisted of a total of 495 feet of tunnelling and 2 inclined shafts which had a total length of 255 feet. The Number 1 Zone was supposedly tested over a length in excess of 5000 feet (1524 metres) on surface, by 30 trenches, and the Number 2 Zone was tested over a length of 250 feet (76 metres) with 4 trenches. The Roseberry occurrence was tested by 895 feet of drifting and crosscutting on 3 levels.

In 1934, T.E. Arnold acquired the claims.

In the early 1940's, Rain dor Gold Mines extended the 986 level adit to a length of 500 feet (152.4 metres), sank 2 shallow shafts and dug several surface trenches.

During the period 1962 to 1967, Westairs Mines Limited carried out an exploration program which consisted of geological mapping, prospecting, trenching, underground exploration development and diamond drilling. The diamond drilling (1004 feet) was completed at the Roseberry and A & E Prospects. A 320 foot crosscut was driven north of the workings and the A & E 6000 foot level adit was driven to 265 feet, to facilitate drilling. They also completed 975 feet (297.18 metres) of underground development on the J & L Prospect, at the 830 metre level. Westairs drilled at least 10 diamond drill underground holes from within the 830 metre level adit.

Pan American Minerals Corporation of Vancouver optioned the claims in late 1980 and 11 claim blocks (158 mineral units). Then, in June of 1981, Arnold had an additional 4 claim blocks (72 mineral units) staked. On December 16, 1981, Pivak Explorco Limited optioned the property. The geology of the area has been studied twice by the G.S.C. In the mid 1920's, H.C. Gunning completed his work while in the early 1960's, J.O. Wheeler did his mapping. Numerous geological studies have been completed to the north of the property area. During July of 1982

Geotext Consultants of Vancouver, under contract for the G.S.C., completed a quick geological survey of the area between Carnes Peak and McKinnon-Carnes Creeks.

The only significant producing mines in the area were the Mastodon Highland Bell (south of the J & L) which produced 6,112 ounces silver, 180,334 pounds lead, 5,911,618 pounds zinc and 24,716 pounds cadmium from 31,900 tons of ore and the Goldstream massive sulphide deposit which was discovered in 1973 and is owned by Noranda Exploration Company Limited. It is approximately 43 km north-northwest of the J & L property and had reserves (diluted) of 3.94 million tonnes grading 3.7% copper, 2.7% zinc and 0.56 ounces/tonne silver. The discovery of this deposit has increased exploration activity in this area in recent years.

Amax, Long Lac Minerals and Union Oil were active in the Downie Creek area during 1982 and 1983. It appears that they have been working on high elevation tungsten prospects. Preussag Canada Limited were also active at the Standard Prospect which is to the northwest of the J & L. A number of independents and junior companies were also in

the region, investigating carbonatite, tungsten, copper, gold and massive sulphide prospects.

During 1982, Selco completed a preliminary geological and geochemical surface survey of the important part of the claim area. This included reconnaissance geology, 1:2 500 grid mapping and detailed mapping and sampling of the known surface showings. Selco had two airborne electromagnetic surveys flown over the property. Road building and repair, bridge construction and line cutting were also done.

During 1983, Selco completed a detailed geological and geochemical surface survey on Goat Mountain. This included detailed mapping and sampling of mineralized showings, 1:100 grid mapping and 1:5 000 reconnaissance mapping. Preliminary mapping (1:5 000) of part of Roseberry Mountain, to the north of the J & L Main Zone, was also completed. The 1982 airborne geophysical anomalies were investigated with ground geophysics (Genie EM and Mag.) and/or reconnaissance geology. An underground geological and geochemical study of part of the J & L Main Zone, at the 830 metre level elevation, was also completed. This consisted of 354.94 metres of track drift which followed

the Main Zone mineralization and 3 hangingwall crosscuts which totalled 216.60 metres. This underground development was an extension of the drifting completed by Westairs Mines Limited during the 1960's. Subsequent drilling consisted of 35 holes which totalled 1657.89 metres.

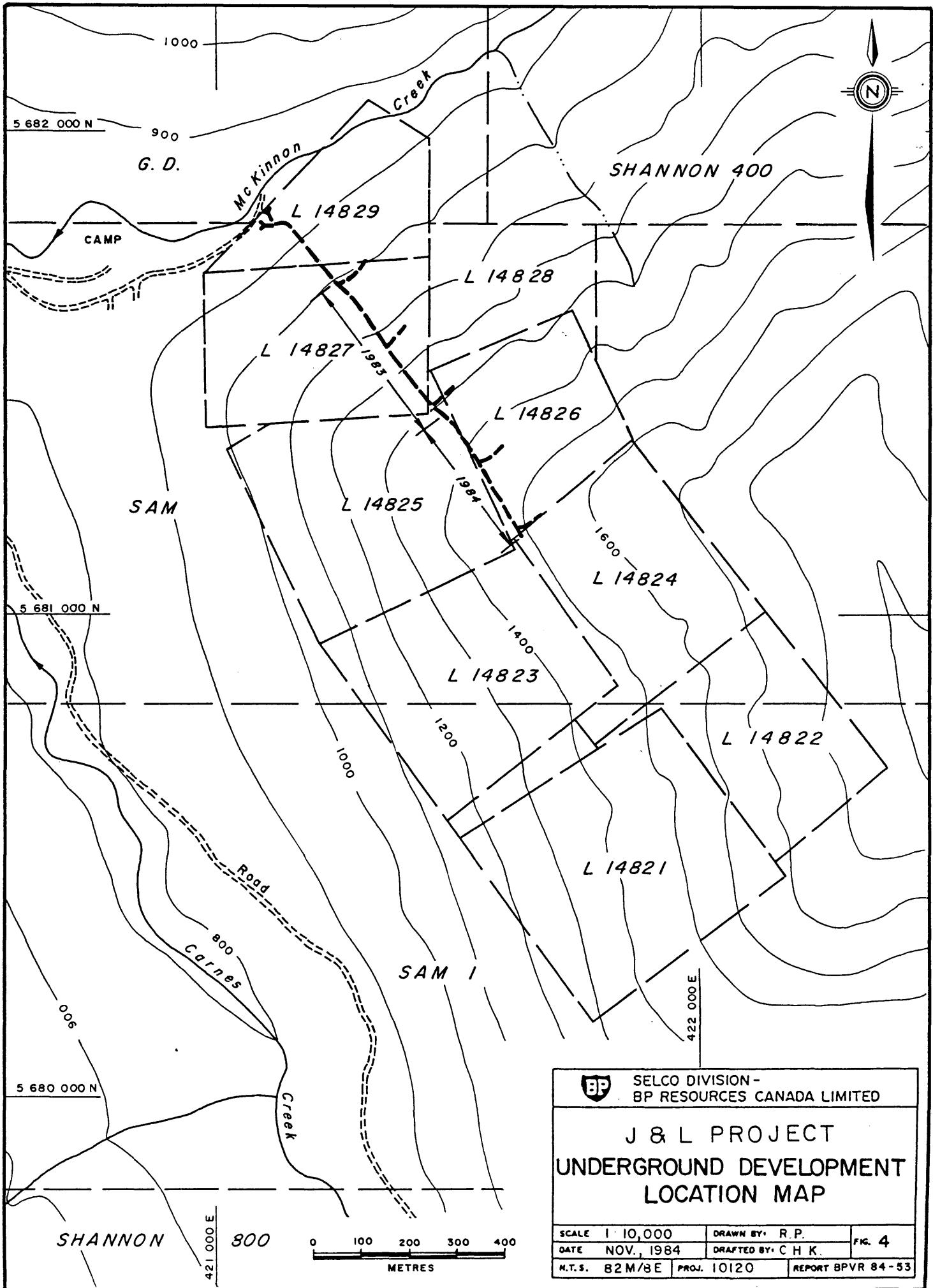
4. 1984 Work Program Summary

Compilation of the J & L data and the organization and planning of the 1984 program continued until late May, when the field work began.

At this time, part of the field crew had completed the rehabilitation of the Selco camp. Due to weather conditions, regional mapping and camp work was done until the third week of June when mapping on Roseberry Mountain, to the north-northwest of the 830 metre level portals, began. The 986 portal area was cleared out at this time. The surface crew completed a chained, flagged and compassed grid on Roseberry Mountain during late June. This crew managed to extend the arsenical mineralization to the northwest and completed detailed mapping along the grid (See Figure 10).

The underground contractor camp was re-established and preliminary road and adit rehabilitation was completed during early July. On July 14th, the underground exploration development began, from the end of the 1983 work. This new development consisted of a track drift (2.1 m x 2.1 m) along the Main Zone (315.80 m) and 2 crosscuts into the hangingwall stratigraphy (136.23 m). Diamond drill stations were established at the ends of the crosscuts, from which up and down holes were drilled. These holes tested the up and down dip extensions of the Main Zone. Additional drilling was done from the drift, in order to outline the lateral extent of the mineralization at the drift level. Two holes were also drilled from the drill station at the end of the 10,500E crosscut. The -90° hole was drilled in order to assess the 1983 trend of increasing grade and/or thickness of the zone with depth. The short, +20° hole into the hangingwall was drilled to test for possible mineralization associated with a limestone bed found at the end of the crosscut. A total of 982.51 metres of drilling was completed in 30 drill holes (See Figure 5).

All of the drill core was logged and the appropriate sections were split. The backs and walls of the new drift



and crosscuts were mapped and surveyed as the development progressed. Each "second round" drift face and the crosscut wall(s) were extensively chip sampled and mapped. A total of 993 underground samples and 291 core samples were sent out to be assayed. Metallurgical samples were collected and specific gravity determinations were also completed.

Road repair, waste and ore disposal and environmental water sampling continued throughout the field season, which ended on October 23rd.

PHYSICAL WORK

During May of 1984, a Selco crew began the rehabilitation of the Selco camp, which is located on the south side of McKinnon Creek (See Figure 4). Selco personnel also completed several trails and a compassed, chained and flagged control grid on the south slope of Roseberry Mountain and re-opened the portal of the 986 metre adit on Goat Mountain, during late June and July. The grid consists of a 1200 metre, approximately, baseline, which was measured at an azimuth of 307° from the original (1983) North Zone Showing (See Figure 10, Showing NS-1) and crosslines which were put in, where possible, at 200 metre intervals. This grid was established in order to give the geological mapping better control.

TABLE I
J & L 1984 SURVEY STATIONS (Underground)

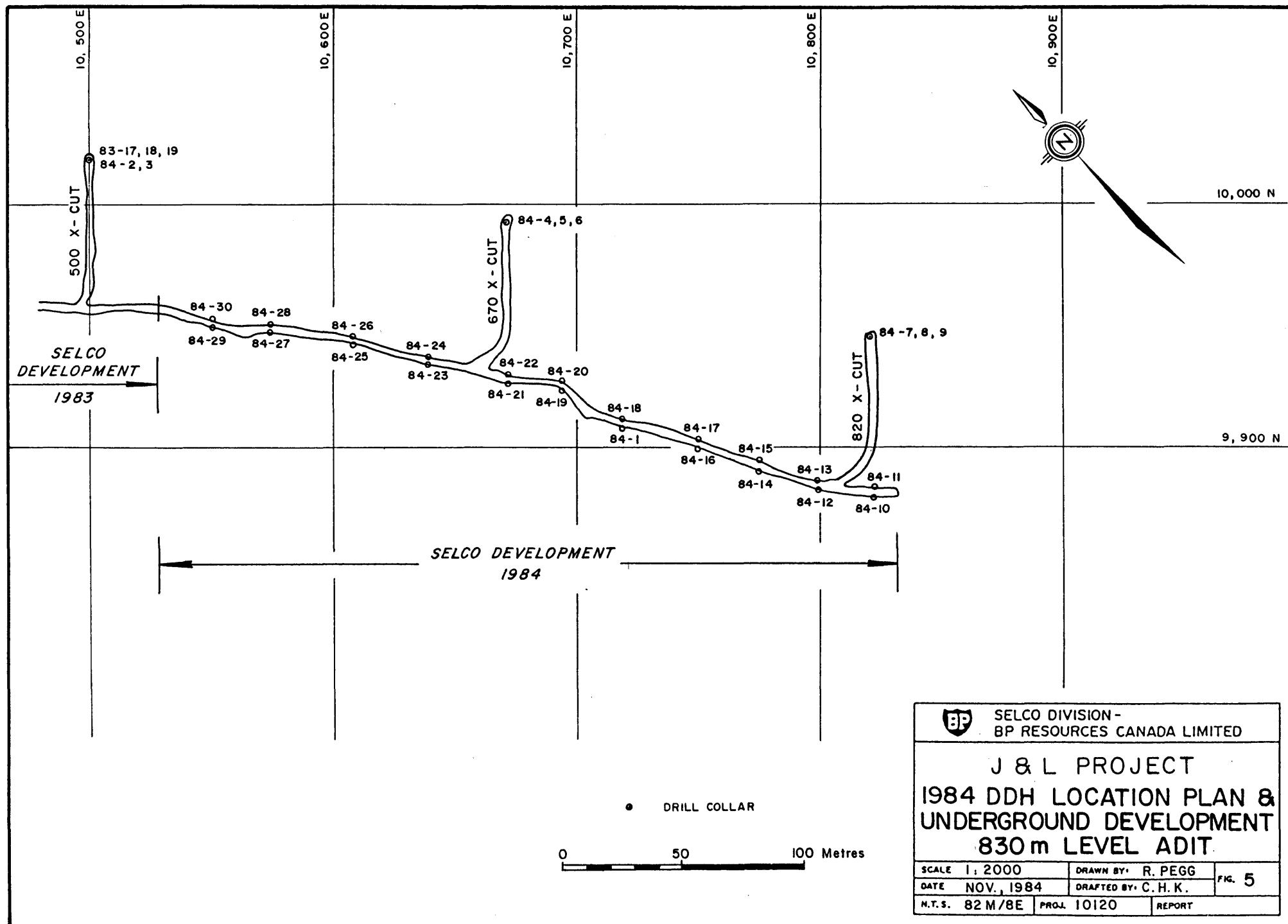
STATION	NORTHING	U.T.M.	EASTING	MINE GRID	EASTING	(m) ELEVATION (R.L.)
J-47	5,681,417.968		421,470.687	9,958.839	10,525.435	839.405 (836.790)
J-47(L.P.)				9,958.114	10,529.171	839.530
J-48	5,681,411.897		421,475.013	9,957.223	10,532.714	839.200 (836.820)
J-48(L.P.)				9,955.895	10,536.452	839.60
J-49	5,681,401.412		421,480.628	9,953.190	10,543.902	839.401 (836.946)
J-49(L.P.)				9,952.348	10,547.024	839.556 (836.942)
J-50	5,681,390.281		421,487.699	9,949.649	10,556.605	839.859 (837.053)
J-50(L.P.)	Destroyed July 25 and replaced with:			9,949.529	10,559.294	839.619 (837.047)
J-50(L.P. ²)				9,949.632	10,559.736	839.675
J-51 ²	5,681,380.673		421,499.847	9,950.637	10,572.062	839.676 (837.033)
J-51(L.P.)				9,949.814	10,575.079	839.665 (836.949)
J-52	5,681,372.155		421,506.075	9,948.475	10,582.390	839.652 (837.056)
J-52(L.P.)				9,947.676	10,586.449	839.944
J-53	5,681,364.110		421,511.983	9,946.457	10,592.157	839.976 (837.183)
J-53(L.P.)				9,946.192	10,594.235	839.941
J-54	5,681,357.349		421,517.984	9,945.448	10,601.139	839.876 (837.222)
J-54(L.P.)	Destroyed & replaced by LP ₂			9,944.150	10,604.011	840.029
J-54(L.P. ²)				9,944.356	10,604.218	840.073 (837.305)
J-55 ²	5,681,348.140		421,522.041	9,941.319	10,610.316	840.066 (837.299)
J-55(L.P.)				9,940.884	10,613.244	840.111 (837.347)
J-56	5,681,337.861		421,530.544	9,939.368	10,623.512	840.049 (837.352)
J-56(L.P.)				9,938.687	10,626.400	840.211
J-57	5,681,328.888		421,537.156	9,937.123	10,634.428	840.125 (837.384)
J-57(L.P.)				9,936.596	10,637.617	
J-58	Destroyed & replaced by J-59			9,933.073	10,653.532	840.237 (837.454)
J-58(L.P.)	Changed to 59 LP			9,932.302	10,656.797	840.192
J-59	5,681,313.254		421,548.535	9,933.117	10,653.345	840.211 (837.481)
J-60	5,581,303.919		421,554.594	9,930.234	10,664.091	840.515 (837.536)
J-60(L.P.)				9,929.461	10,666.378	840.186
J-61	5,681,310.342		421,561.076	9,939.345	10,664.616	840.244 (837.571)

TABLE 1
(continued)

STATION	NORTHING	<u>U.T.M.</u>	EASTING	NORTHING	<u>MINE GRID</u>	EASTING	ELEVATION (m) (R.L.)
(670 X-C)							
J-61(L.P.)							
J-62	5,681,292.992		421,560.396	9,925.995		10,675.718	840.519 (837.551)
J-62(L.P.)				9,926.125		10,679.136	840.372
J-63	5,681,313.273		421,570.078	9,947.546		10,669.344	840.291 (837.616)
(670 X-C)							
J-63(L.P.)	(63 LP is the B.S.)			9,944.245		10,669.343	840.400
(670 X-C)							
J-64	5,681,284.030		421,570.946	9,926.394		10,689.555	840.623 (837.561)
J-64(L.P.)				9,925.273		10,692.329	840.212
J-65	5,681,279.419		421,574.051	9,925.045		10,694.948	840.462 (837.596)
J-65(L.P.)				9,923.424		10,696.293	840.692
J-66	5,681,272.186		421,573.870	9,919.549		10,699.653	840.377 (837.678)
J-66(L.P.)				9,917.851		10,702.562	840.585
J-67	5,681,325.038		421,580.670	9,963.377		10,669.344	840.403 (837.605)
(670 X-C)							
J-67(L.P.)				9,966.731		10,669.344	840.467
J-68	5,681,255.873		421,580.169	9,911.640		10,715.249	840.266 (837.506)
J-68(L.P.)							(Raised) Sep.8 (837.692)
J-69	5,681,342.799		421,597.519	9,911.237		10,717.422	840.240
(670 X-C)				9,987.850		10,669.982	840.472 (837.603)
J-70	5,681,240.573		421,591.564	9,907.894		10,733.955	840.654 (837.822)
J-70(L.P.)				9,906.831		10,736.863	
J-71	5,681,233.894		421,595.172	9,905.345		10,741.105	840.625 (837.931)
J-71(L.P.)				9,904.248		10,743.522	
J-72	5,681,215.056		421,603.370	9,896.830		10,759.802	840.802 (838.046)
J-72(L.P.)				9,896.301		10,761.597	840.797
J-73	5,681,197.750		421,613.828	9,890.966		10,779.154	840.965 (838.185)
J-73(L.P.)				9,890.150		10,781.434	840.785

TABLE 1
(continued)

STATION	NORTHING	U.T.M.	EASTING	MINE GRID		(m) ELEVATION (R.L.)
				NORTHING	EASTING	
J-74	5,681,182.459		421,622.530	9,885.423	10,795.852	841.101 (838.223)
J-74(L.P.)				9,884.757	10,798.303	
J-75	5,681,175.268		421,627.479	9,883.390	10,804.342	841.366 (838.248)
J-75(L.P.)	(820 X-C)			9,884.783	10,806.417	841.212
J-76	5,681,173.901		421,641.335	9,891.645	10,815.554	840.992 (838.210)
(820 X-C)						
J-76(L.P.)	(820 X-C)			9,893.227	10,816.247	841.064
J-77	5,681,168.346		421,631.824	9,881.153	10,812.203	841.033 (838.242)
J-77(L.P.)				9,881.454	10,815.737	841.027
J-78	(820 X-C) Destroyed & Replaced by J79			9,898.843	10,819.774	840.708
J-78	(back) L.P.					
J-79	5,681,176.436		421,649.318	9,898.870	10,819.791	840.775
(820 X-C)						
J-79	(820 X-C)			9,901.403	10,819.783	841.231
J-80	5,681,159.646		421,641.879	9,881.419	10,825.497	841.049 (838.349)
(Final Station - Main Zone)						
J-81	5,681,185.892		421,657.832	9,911.594	10,819.792	841.269 (838.366)
(820 X-C)						
J-81(L.P.)				9,913.842	10,819.791	841.089
J-82	5,681,199.488		421,670.082	9,929.894	10,819.798	841.236 (838.460)
(820 X-C)						
J-82(L.P.)				9,933.022	10,819.799	
J-83	5,681,210.983		421,680.783	9,945.597	10,820.059	842.221 (838.497)
J-83(L.P.)				9,944.033	10,820.103	



During the period of July 6 to July 13, Rocbore (1980) Limited of Kamloops, completed their crew and equipment mobilization and rehabilitation of the J & L access road, the 830 metre adit portal area and the previous 830 metre level underground development.

Rocbore then extended the 830 metre level, track drift, which measures 2.1 x 2.1 metres, for approximately 315.8 metres. The extention of the drift was again driven under geological control in an effort to follow the Main Zone mineralization. The two new crosscuts, which were under survey control, were driven into the hangingwall stratigraphy. The survey controlled production, which totalled 136.23 metres, included 68.40 metres in the 10,670E crosscut and 67.83 metres in the 10,820E crosscut. A total of 132.02 cubic metres of slashing was also completed and this consisted of 42.30 cubic metres within the drift, 53.27 cubic metres within the 10,670E crosscut and 36.45 cubic metres within the 10,820E crosscut. The slashing included the creation of diamond drill stations at the ends of the crosscuts. A minor amount of rock bolting was required in the drift and crosscuts. This underground development program was completed during the period of July 14 to October 7, 1984.

TABLE II
J & L D.D.H. COORDINATES

DRILL HOLE	MINE GRID U.T.M.		(m) ELEV. (RAIL)	AZIM.	DIP	(m) LENGTH	HOLE TYPE	DATE STARTED	DATE COMPLETED
	NORTHING	EASTING							
84-1	9,909.950 5,681,252.480	10,718.442 421,581.411	838.929	222°	+0.96°	27.74	Definition Drilling, Drift	Sep. 1/84	Sep. 2/84
84-2	10,015.256 5,681,476.764	10,500.224 421,489.701	837.054	222°	-87.75°	97.54	10,500E Crosscut	Sep. 4/84	Sep. 7/84
84-3	10,018.576 5,681,479.187	10,500.290 421,491.972	838.800	042°	+19.69°	15.24	10,500E Crosscut	Sep. 7/84	Sep. 7/84
84-4	9,990.536 5,681,344.939	10,669.767 421,599.156	837.495	222°	-89.5°	119.48	10,670E Crosscut	Sep. 8/84 Sep. 16/84	Sep. 10/84 Sep. 17/84
84-5	9,989.563 5,681,344.235	10,669.738 421,598.484	837.527	222°	-45.5°	90.53	10,670E Crosscut	Sep. 10/84	Sep. 13/84
84-6	9,989.117 5,681,343.906	10,669.776 421,598.239	840.506	222°	+18.25°	112.47	10,670E Crosscut	Sep. 13/84	Sep. 16/84
84-7	9,942.830 5,681,209.168	10,819.698 421,678.663	838.463	222°	-44.4°	70.10	10,820E Crosscut	Oct. 8/84	Oct. 9/84
84-8	9,943.711 5,681,209.824	10,819.697 421,679.252	838.473	222°	-87.3°	109.12	10,820E Crosscut	Oct. 10/84	Oct. 11/84
84-9	9,942.453 5,681,208.845	10,819.762 421,678.459	841.381	222°	+17.97°	109.12	10,820E Crosscut	Oct. 13/84	Oct. 14/84
84-10	9,880.312 5,681,160.863	10,822.448 421,638.872	839.401 (838.356)	222.36°	-0.4°	10.06	Definition Drilling, Drift	Oct. 15/84	Oct. 15/85

TABLE II
(continued)

DRILL HOLE	MINE GRID U.T.M.		(m) ELEV. (RAIL)	AZIM.	DIP	(m) LENGTH	HOLE TYPE	DATE STARTED	DATE COMPLETED
	NORTHING	EASTING							
84-11	9,882.900 5,681,162.764	10,822.482 421,640.629	839.404	042.7°	-0.2°	12.49	Definition Drilling, Drift	Oct.15/84	Oct.15/84
84-12	9,983.273 5,681,178.228	10,799.788 421,624.016	839.531 (838.284)	222.4°	+0.4°	10.06	Definition Drilling, Drift	Oct.15/84	Oct.15/84
84-13	9,886.505 5,681,180.838	10,799.477 421,625.948	839.493	042.51°	-0.5°	10.05	Definition Drilling, Drift	Oct.15/84	Oct.15/84
84-14	9,891.049 5,681,200.428	10,775.245 421,610.978	839.392 (838.174)	222.4°	-0.1°	9.75	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-15	9,893.644 5,681,202.603	10,774.875 421,612.439	839.370	040.9°	-0.2°	10.06	Definition Drilling, Drift	Oct.15/84	Oct.16/84
84-16	9,899.329 5,681,223.633	10,749.764 421,597.596	839.304 (838.069)	222°	-0.2°	10.06	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-17	9,902.742 5,681,226.391	10,749.427 421,599.615	839.352	042.5°	+1.6°	10.06	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-18	9,912.719 5,681,254.677	10,718.235 421,583.110	839.020 (837.782)	044.1°	+0.6°	9.75	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-19	9,923.643 5,681,278.756	10,694.394 421,572.709	838.850 (837.666)	222.7°	-0.3°	10.06	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-20	9,926.490 5,681,281.057	10,694.106 421,574.392	838.856	043.7°	-0.3°	10.06	Definition Drilling, Drift	Oct.16/84	Oct.16/84
84-21	9,926.334	10,671.630	838.763	223.5°	-1.1°	9.75	Definition		

TABLE II
(continued)

DRILL HOLE	MINE GRID U.T.M.		(m) ELEV. (RAIL)	AZIM.	DIP	(m) LENGTH	HOLE TYPE	DATE STARTED	DATE COMPLETED
	NORTHING	EASTING							
84-22	9,929.151 5,681,298.256	10,671.356 421,559.266	838.794	043.4°	+0.3°	9.14	Definition Drilling, Drift	Oct.17/84	Oct.17/84
84-23	9,935.580 5,681,325.493	10,637.788 421,538.620	838.675 (837.472)	222.3°	+0.2°	13.11	Definition Drilling, Drift	Oct.17/84	Oct.17/84
84-24	9,938.012 5,681,327.299	10,637.790 421,540.250	838.691	042.0°	+0.8°	9.75	Definition Drilling, Drift	Oct.17/84	Oct.17/84
84-25	9,941.401 5,681,348.140	10,607.987 421,520.365	838.543 (837.291)	223.4°	+0.3°	11.58	Definition Drilling, Drift	Oct.17/84	Oct.17/84
84-26	9,944.448 5,681,352.202	10,607.720 421,522.205	838.551	044.1°	+0.3°	12.50	Definition Drilling, Drift	Oct.17/84	Oct.17/84
84-27	9,948.551 5,681,377.514	10,574.467 421,500.239	838.272 (837.080)	222°	-0.3°	17.68	Definition Drilling, Drift	Oct.18/84	Oct.18/84
84-28	9,951.418 5,681,379.863	10,574.140 421,501.915	838.255	042.2°	0.0°	9.60	Definition Drilling, Drift	Oct.18/84	Oct.18/84
84-29	9,949.332 5,681,394.053	10,550.615 421,483.036	838.268 (838.007)	221.7°	+0.4°	15.85	Definition Drilling, Drift	Oct.18/84	Oct.18/84
84-30	9,952.544 5,681,396.446	10,550.291 421,484.753	838.257	043.2°	-0.1°	9.75	Definition Drilling, Drift	Oct.18/84	Oct.18/84

1984 DRILLING SUMMARY

1. 10,500E Crosscut Drilling (2 holes) = 112.78 m
2. 10,670E Crosscut Drilling (3 holes) = 322.48 m
3. 10,820E Crosscut Drilling (3 holes) = 288.34 m
4. J.V. Drilling (22 holes) = 258.91 m

TOTAL DRILLING = 982.51 m

The 1984 underground surveying (See Table I) was done by the Selco site engineer who continued to utilize the mine grid established in 1983. This mine grid, whose ON, OE coordinates are found at 5,681,060N and 406,985E (U.T.M.) was established with "mine north" at an azimuth of 042°.

Between September 1 to 17 and October 8 to 18, 1984, Connors Drilling Ltd. of Kamloops, who were subcontracting for Rocbore, completed the underground drilling, which totalled 982.51 metres of BQ core. This drilling, which was done with an electric drill, consisted of 258.91 metres of core from 22 holes ("J.V.") collared within the new drift development, 112.78 metres of core from 2 holes at the end of the 10,500E crosscut, 322.48 metres of core from 3 holes at the end of the 10,670E crosscut and 288.34 metres of core from 3 holes at the end of the 10,820 crosscut (See Table II). All of the crosscut drill holes were surveyed with Sperry Sun apparatus by Selco personnel.

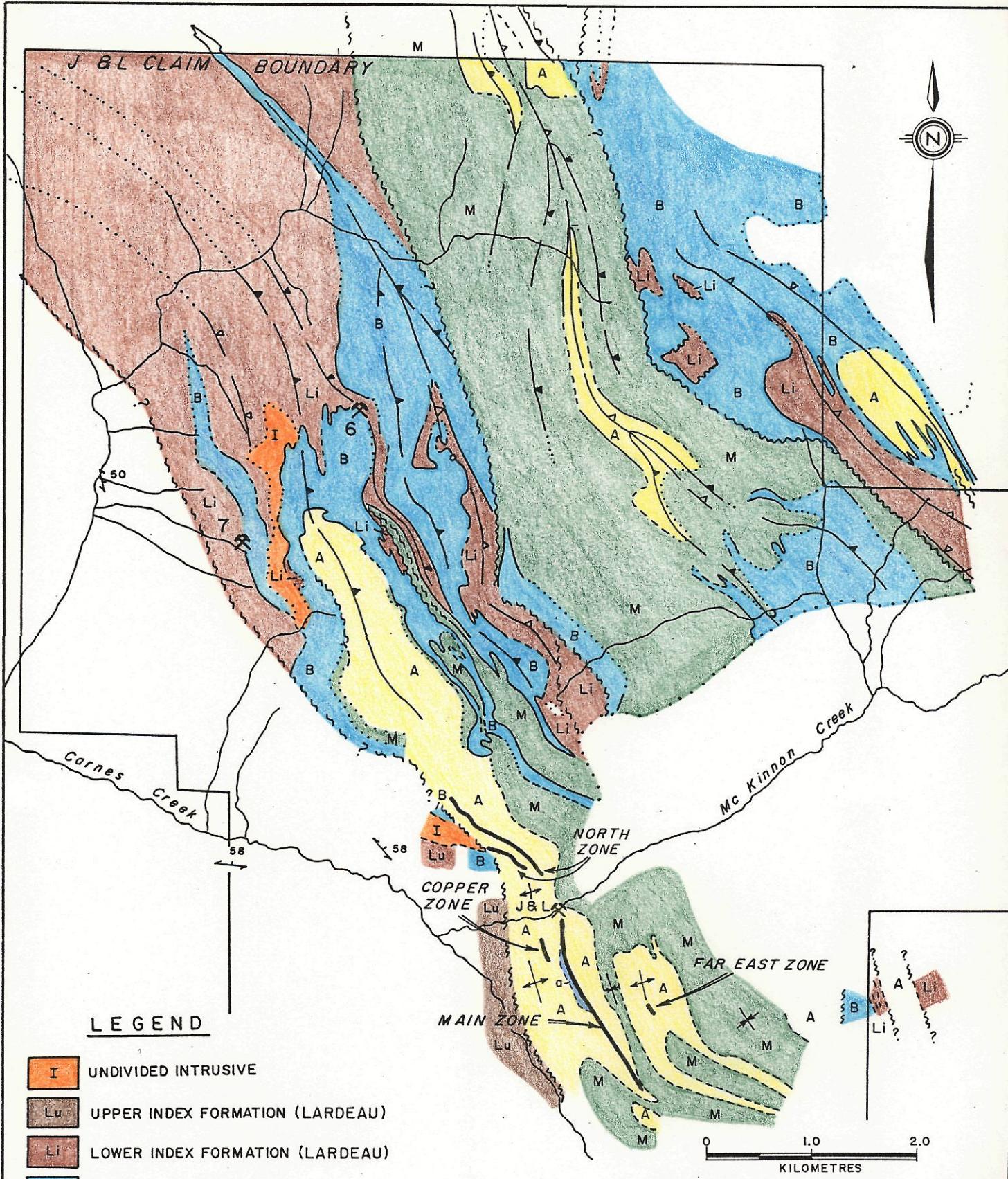
GEOLOGY

1. Regional Geology

The G.S.C. (Wheeler, 1963) has mapped the J & L Project area within a northwest trending belt of metamorphic rocks of the Lardeau Group (Lower Cambrian and later). This belt

is bordered on the west by the Columbia River fault and the Shuswap Metamorphic Complex, on the south by a migmatite complex of the Shuswap Metamorphic Complex, on the north (approximately) by the Goldstream River and on the east by Downie Creek and sediments of the Hamill Group and the Badshot Formation. The Lardeau Group is comprised of slates, phyllitic siltstones, quartzites, schists and limestones. Wheeler has also indicated the presence of numerous discontinuous beds of limestone, dolomite, phyllite and quartzite of the Badshot Formation (Lower Cambrian), several large granitic bodies (Post Lower Cambrian) and 2 small pods of Hamill Group (Lower Cambrian) quartzite within this large block of folded and faulted Lardeau rocks (See G.S.C. Map 12-1964).

During 1976 and 1977, Trygve Hoy of the B.C.D.M. mapped and compiled the geology for the Goldstream area and he postulated that the stratigraphy consists of Hamill Group metasediments and metavolcanics and minor Horsethief Creek Group (Lower Paleozoic-Upper Proterozoic) metasediments and metavolcanics (to the east) which have been intruded by several granitic bodies. These metasediments and metavolcanics are intensely deformed and consist of



SELCO DIVISION -
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J & L PROJECT REGIONAL GEOLOGY



SCALE 1:50,000

DRAWN BY: R. PEGG

DATE JAN. 1985

DRAFTED BY: L.G. + J.S.

FIG. 6

N.T.S. 82 M / 8E PROJ. 10120

REPORT BPVR 84-53

quartzites, schists, phyllites, calc-schists and carbonates, which are interlayered with greenstones and chloritic phyllites. Tight to isoclinal north-trending folds are the dominant structure in the area. The folds are overturned with their axial surfaces varying from nearly horizontal in the Keystone area to steeply dipping in the Downie and Standard areas (See Plan 1).

Mapping by Geotex, consultants for the G.S.C. in 1982, indicates that the stratigraphy, southeast of Burke Creek, consists of folded and faulted Lardeau, Mohican and Hamill metasediments and metavolcanics and Badshot limestones (fossiliferous on Roseberry Mountain). Geotex stated that "the distribution of stratigraphic boundaries results from the interference between two phases of non-coaxial isoclinal folding and subsequent faulting". The Geotex mapping on Roseberry Mountain has been, in part, confirmed by Selco geologists. Geotex also postulated that the area around Standard Peak is comprised of metamorphics of the Lardeau Group and not the Hamill, as Hoy had indicated.

It appears that J & L Main Zone on Goat Mountain is found within a belt of Hamill Group phyllitic quartzites whose

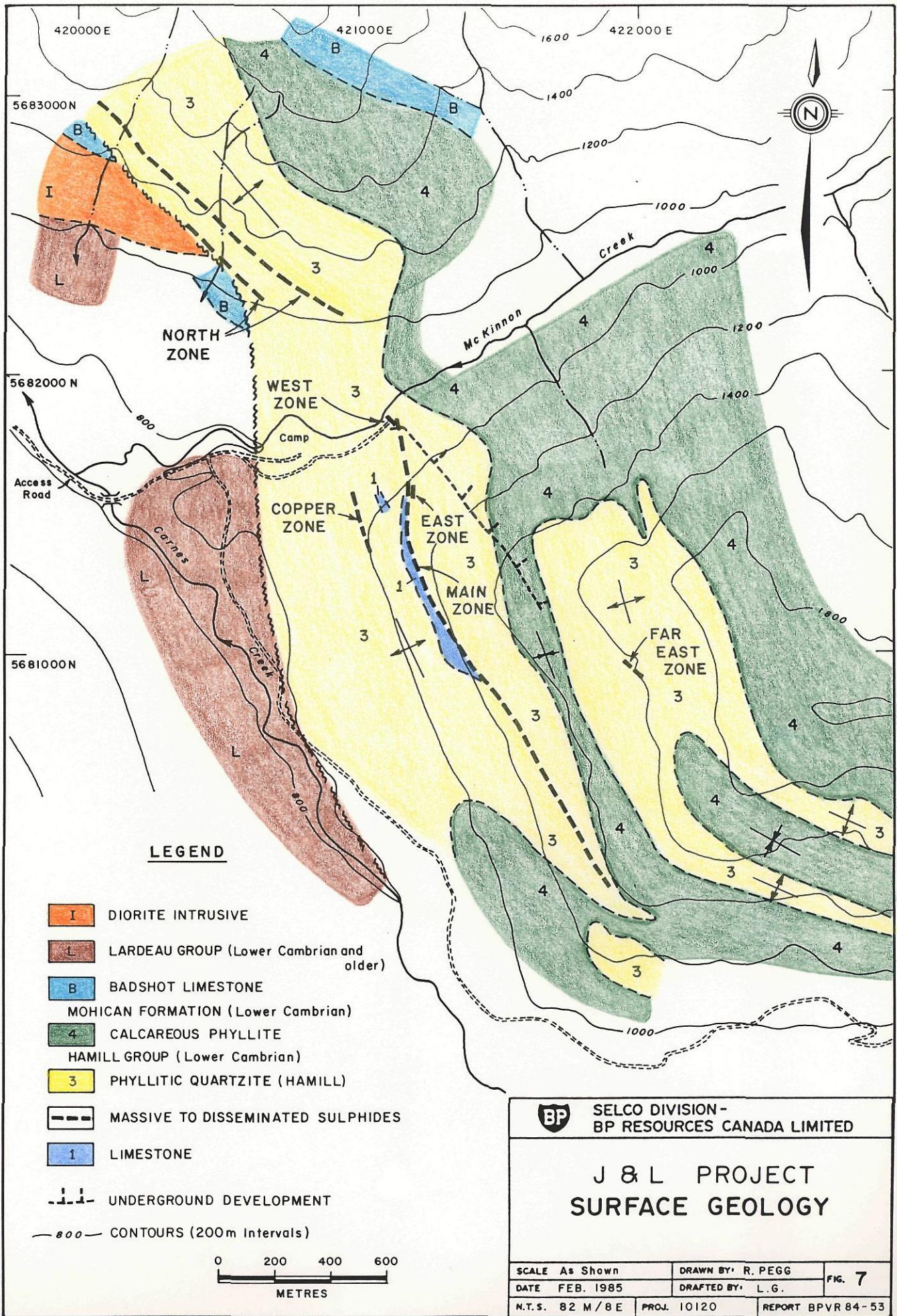
eastern limb is comprised of calcareous phyllites and limestones of the Mohican Formation. A northwest trending thrust fault separates the younger Lardeau phyllites on the west and the Hamill Group metasediments on the east. The Hamill Group also contains a number of discontinuous beds of grey banded limestone, the most prominent of which forms the footwall to the Main Zone for approximately 900 metres.

Wheeler postulated that the metamorphic belt, which may include the Hamill Group, extends south of the J & L property to La Forme Creek.

2. Surface Geology

Surface mapping concentrated on extending the arsenical North Zone to the northwest of the 1983 discovery occurrence (NS-1, See Figure 11) and completing a detailed (1:100) geological study of the stratigraphy on the southwest slope of Roseberry Mountain (See Plan 2).

North to northwest striking, moderate to steep, east dipping sediments of the Hamill, Mohican, Badshot and Lardeau Formations cover most of the investigated area on Roseberry Mountain. Generally, isoclinally folded,



phyllitic quartzites of the Hamill Group appear to form the core of an antiform whose southwestern flank has been disrupted by a multiphase, late stage, dioritic intrusive. Interfingered graphitic phyllites of the Lardeau Group and limestones of the Badshot Formation were observed as remnants within the intrusive and below its southern margin. The northeastern flank of this quartzitic antiform core is overlain by limestones and dolomites of the Mohican Formation which are, in turn, overlain by limestones of the Badshot Formation.

The host of the North Zone mineralization is the phyllitic quartzite unit of the Hamill Group. This unit consists of intercalated clean to dirty quartzites and quartz-sericite to sericite phyllites. A few narrow and discontinuous grey banded limestone beds were also noted within the Hamill Group.

The overlying Mohican Formation is a calcareous phyllite unit which is comprised of limestones, tan weathering dolomites, calcareous grits and phyllites and minor calcareous quartzites. Overlying the Mohican Formation, to the east, are grey banded limestones of the Badshot Formation.

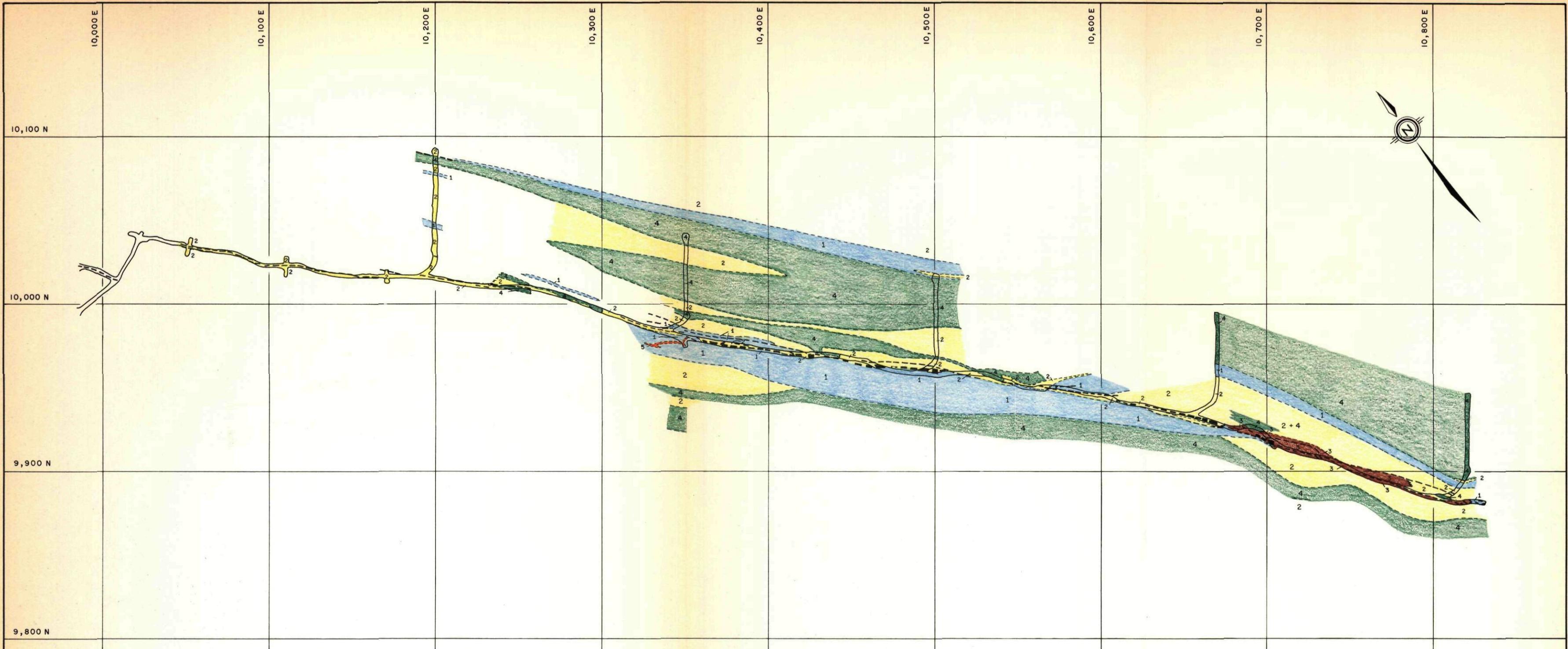
On the southwest side of the grid, the Hamill Group has been in part cut off and thrust up against a mix of Lardeau and Badshot metasediments and an apparently younger diorite sill complex. The dioritic complex appears to be sill-like with observed local conformable contacts with the Lardeau metasediments. The sill reaches a maximum true thickness of 150 metres, but splits into thinner coalescing sill-like bodies, along strike. The sills thicken and thin along strike and have differentially penetrated the surrounding metasediments. The diorite is a multiphase intrusive with a highly variable mineralogical composition. It ranges from a black biotite-rich, fine-grained rock to a hornblende or plagioclase-quartz porphyry to an amphibolite, all with variable carbonate contents. Zonation within the intrusive is fairly vague, except for the biotite-rich unit which is found within the core of the thicker sections. The intrusive is, locally, well foliated with foliations parallel to the regional foliations and has pervasively undergone chloritic alteration. This sill appears to be similar to the one that was mapped by Geotex, further to the north on Roseberry Mountain.

The Badshot Formation consists of small, discontinuous limestone horizons which have been disrupted by the intrusive and interfingered with the dirty clastics of the Lardeau Group. The Lardeau Group consists of a thick succession of graphitic phyllites, sericitic phyllites and chloritic quartzites and phyllites. The emplacement of the diorite appears to have resulted in local veining, brecciation and folding within the Lardeau stratigraphy.

3. Underground Geology*

The 1984 underground development and drill program exposed an intercalated sequence of deformed sediments and metasediments of the Hamill Group which host the J & L Main Zone. This sequence consists of quartzites, quartz-sericite to sericite phyllites, argillaceous metagreywackes, limestones and carbonaceous limestones (See Figure 8 and Plan 4).

*Please note that when referring to the underground development, orientations are based on the mine grid.



LEGEND

- [5] LAMPROPHYRE DYKE
- [4] CHLORITIC PHYLLITES
- [3] SERICITIC PHYLLITES
- [2] PHYLLITIC QUARTZITES
- [1] MASSIVE SULPHIDES
- [1] LIMESTONE

0 20 40 60 80 100
METRES

BP SELCO DIVISION -
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J & L PROJECT
SIMPLIFIED GEOLOGY
830 m LEVEL DRIFT

SCALE 1:2,000	DRAWN BY: R. PEGG	FIG. 8
DATE JANUARY 1985	DRAFTED BY: L.G.	
N.T.S. 82 M / 8 E	PROJ. 10120	REPORT BPVR 84-53

Generally, the quartzites and quartz-sericite to sericite-quartz phyllites occur as the host and immediate structural hangingwall of the Main Zone mineralization. The carbonaceous to grey banded limestone is found as the structural footwall to the Main Zone, until, approximately, 10704E where it pinches out. East of this point, a sericite-quartz to quartz-sericite to sericite phyllite occurs in the drift as the immediate structural footwall and hangingwall until, approximately, 10825E where the carbonaceous limestone reappears, as the host rock. Local lenses and discontinuous narrow sequences of argillaceous metagreywackes were also noted at or near the immediate hangingwall. It should be noted that between 10761.87E and 10787.7E, a large, mineralized, coarse-grained, white quartz body occurs as part of the Main Zone. To the east of this quartz body, the immediate structural hangingwall shows a pronounced, visible increase in silica content.

Generally, the structural footwall, beneath the limestone and east of where it pinches out, consists of an intercalated sequence of quartzites, siliceous phyllites and argillaceous metagreywackes. The hangingwall

stratigraphy is comprised of an intercalated sequence similar to the footwall. The 10670E crosscut revealed a thick hangingwall sequence of siliceous metasediments which thins substantially towards the 10820E crosscut. A thin (2-6 metre) carbonaceous to grey banded limestone band was also observed and this maybe part of a folded limestone-sulphide sequence found at 10570E (See Figure 8) or a separate discontinuous limestone band. The definition drilling from the 830 metre drift did not extend far enough into the hangingwall stratigraphy to fully determine the character of this limestone (See Plan 4). It should also be noted that individual argillaceous and siliceous metasediment units appear to be discontinuous and can grade into one another along strike.

The quartzite (06) group consists of clean to dirty, creamy-white quartzites and green quartz-chlorite phyllites. The quartzites are fine-grained, massive to weakly banded and carry a minor sericitic component along the foliation planes. The quartz-chlorite phyllites are moderately to well foliated and carry a minor chloritic component along the foliation planes.

Closely associated with the quartzites are the quartz-sericite phyllites (07) which are fine-grained, thinly banded, moderately sheared, well foliated and light grey to green in colour. The micaceous component is usually found in distinct sericite-rich bands and can comprise up to 49% of the rock.

The argillaceous metagreywackes (05) were mapped as chlorite, chlorite-sericite and chlorite-quartz phyllites. These fine-grained rocks vary in colour from medium to dark green and are well sheared and foliated. Generally, the micaceous minerals are dominant, but the quartz content can vary from 10 to 49%. It should be remembered that only a small portion of the micaceous minerals are actually chlorite, but it is assumed that its' presence gives this unit its' green colour.

The sericite phyllite unit (04), which is found in close association with the Main Zone mineralization, consists of sericite and sericite-quartz phyllites. These fine-grained rocks are very light green in colour, well sheared and foliated and are very friable locally. The quartz component, which can comprise up to 49% of the rock, is usually found as lenses or discontinuous bands.

The mineralized coarse-grained white quartz unit (13) is approximately 26 metres long and 1.6 metres wide. It is presumed that this quartz body is a segregation of silica from the Main Zone and its' immediate hangingwall and footwall stratigraphy. It was noted during the course of the development that the silica content of the immediate hangingwall and footwall stratigraphies had noticeably decreased east of the 10670E crosscut (See Plan 4). It appears possible that the silica has been concentrated into this quartz "lense" found within the 830 metre drift.

The carbonaceous limestones (02) are generally fine to medium-grained, dark grey to black in colour and poorly to well banded. The dark colour is due to a large argillaceous and graphitic component. Locally, these limestones are highly siliceous (10825E) and can contain up to 50% white carbonate fracture filling.

The grey banded limestones (03) are medium-grained, moderately to well banded, recrystallized rocks. Generally, this sediment consists of a mosaic of carbonate and carbonate cement with distinct thin bands of subordinate argillaceous and clastic components. Locally, this unit is very siliceous (10830E).

4. Underground Mineralization

The Main Zone, which has an approximate average strike of 143° (132° - 150°) and a dip of between 45° - 65° E, consists, essentially, of bands, lenses, stringers and disseminations of massive sulphides. These massive sulphides are comprised of arsenopyrite and pyrite with lesser amounts of sphalerite and galena. Other visible minerals, which are generally seen in minor to trace amounts, include chalcopyrite, pyrrhotite, tetratedrite and a series of lead-antimony sulphosalts. The Main Zone is found up to 3 metres wide in the underground development and includes up to 4 massive sulphide bands with intervening disseminated sections. Complex isoclinal folding has resulted in relative thickening and thinning of sulphide bands along strike. Individual sulphide bands also frequently split into multiple, discontinuous bands which may rejoin to form a continuous band or pinch out over a very short distance. In 1984, a general increase in the width and quantity of sulphides and their grain size is evident.

The 1984 portion of the 830 metre drift continued to follow the Main Zone mineralization with fairly good success. An attempt was made to follow the portion of the zone closest

to the footwall limestones which proved to be the correct course of action.

Along strike, the Main Zone can be separated into 5 separate sections. The first section contains a thick mineralized zone that has the limestone footwall and this extends to 10704E where the limestone pinches out. Between 10704E and 10730E, the massive sulphides have pinched down to very narrow band(s) or lenses and disseminations. This section, generally, contains a very low base metal content. The third section, from 10730-10762E thickens up and there is an appreciable increase in sphalerite and galena. From 10762E to 10788E, the large, coarse-grained mineralized quartz body separates two bands of massive sulphides. Coarse-grained fracture fillings of galena, sphalerite and chalcopyrite with lesser amounts of pyrrhotite and pyrite are found near the outer margins of the quartz. The central core of the quartz is relatively, poorly mineralized. Although the mineralization in this region is erratic, it appears that there is a general increase in the sphalerite and galena content and a decrease in arsenopyrite. East of where the quartz body has terminated, the massive sulphides again thicken. Here, the

sphalerite is abundant, but the galena content has decreased and the arsenopyrite content remains relatively low. The last section appears to be found at the last face where the host limestone has been saturated with extremely abundant, red to honey, very fine-grained sphalerite, lesser amounts of galena and pyrite and very little arsenopyrite.

Lateral sulphide zonations within the Main Zone are very vague, but generally the hangingwall sulphides tend to be more arsenical and those in the footwall, more pyritic. The footwall sulphide bands usually have a poorly to moderately developed banded texture, while the hangingwall band usually displays a "milled" texture with a coarse-grained arsenical and/or pyritic margin(s). This "milled" texture consists of fine to coarse-grained, rounded to subrounded pyrite, occasionally arsenopyrite or sphalerite in a very fine-grained sulphide matrix. The "milling" has been interpreted as a mylonitic texture which has been developed by episodes of strong shearing.

Things of note include irregular patches of extremely coarse-grained arsenopyrite which is found at 10604E and

the thickest massive sulphide section which was observed between 10661E and 10674E. Several narrow, discontinuous pyrrhotite-rich bands were observed at 10591-10662E, 10649-10653E, 10673E and within the folded hangingwall limestone at 10570E. Fine-grained Pb-Sb sulphosalts were observed in minor amounts of several locations, but it is evident that when they are intermixed with the arsenopyrite and/or galena, they would be extremely difficult to recognize. Lenses and discontinuous bands of wallrock were also observed within the massive sulphides.

Mineralization within the footwall stratigraphy consists of minor wispy sphalerite bands, stringers and disseminations within the limestones and minor disseminations and stringers of pyrite, arsenopyrite and sphalerite within the phyllites.

Mineralization within the hangingwall stratigraphy consists of pervasive pyrrhotite along the foliation planes in the chloritic phyllites and minor, thin and discontinuous monomineralic bands, stringers and disseminations of pyrite, arsenopyrite and sphalerite within the quartz-sericite and sericite phyllites. In the 10820E crosscut,

there are also several thin, discontinuous bands and lenses of sphalerite-rich massive sulphides within a sheared sericite-quartz phyllite. Similar sulphides are found at the footwall of the hangingwall limestone band as well as minor disseminated galena and sphalerite within the limestone itself. The hangingwall stratigraphy exposed in the 10670E crosscut is relatively barren of significant mineralization.

5. Structure and Alteration

Dr. Ken McClay of the University of London, England and Brian Marten of Selco, conducted a detailed structural evaluation of the J & L deposit during the fall of 1984.

Dr. McClay's work is summarized below:

The deformational history of the J & L deposit has been long and complex. Five phases of deformation were recognized, of which D3 is the most significant. The F3 folds are near isoclinal, oriented NW-SE with axial planes dipping moderately 45° - 60° to the east. A penetrative S3 foliation has transposed earlier fabrics. Intense shearing along F3 fold limbs has produced tectonic slides which

divide the area into tectonically bounded easterly dipping slices. The J & L massive sulphide sheet occupies one of these shear zones on the limb of an F3 fold. The extreme structural complexity at the J & L deposit precludes any accurate determination of stratigraphic thicknesses or relationships.

McClay's structural mapping enabled him to assemble the following deformation history:

- a) D1 Deformation: This has been recognized in the footwall and hangingwall rocks as small local folds and as a grain fabric (S1) in the quartzitic units of the Hamill Group. The S1 fabric is folded and crosscut by later fabrics. The D1 folds are oriented NW-SE subparallel to the regional structural grain, but this orientation may in part result from transposition of D1 structures by the extremely strong and penetrative D3 deformation. The F1 folding which produced early isoclinal-near isoclinal folds with the development of an intense grain shape fabric S1 in the quartzitic units.

- b) D2 Deformation: This has been recognized by the occurrence of small buckle folds and the development of a spaced pressure solution cleavage S2. The S2 fabric is best observed in the quartzites and can also be seen in thin section as a preferred orientation of sericite flakes. The S2 fabric can clearly be seen to be cutting the compositional banding (S0) and in places is oriented E-W suggesting that some F2 folds were at least locally oriented E-W. The S2 fabric is penetrative and spaced which indicates a component of pressure solution and is folded into crenulation like folds of the D3 deformation.
- c) D3 Deformation: This is responsible for the major folds and the dominant penetrative S3 foliation in the vicinity of the J & L deposit. The major folds are near isoclinal and trend NW-SE with variable plunges from 30° - 350° to 40° - 050° . A penetrative axial planar foliation S3 is developed in these folds. At the J & L, major fold closures are not observed, but asymmetric minor folds reveal the major fold structures.

Minor F3 folds are well developed showing the development of a spaced penetrative axial planar cleavage. In the quartz-sericite phyllites, transportation into this penetrative S3 foliation has obliterated most of the earlier fabrics. In thin section domainal fabrics are seen with the S2 sericite fabric preserved in lensoidal domains.

In addition to the development of minor folds on major F3 fold limbs, ductile D3 shear zones are also developed with quartz vein tension gashes. Quartz veining parallel to S3 or sub-parallel to S3 is common.

A well developed intersection/rodding lineation L3 is found throughout the area mapped, plunging from 30°-345° to 40°-060°. This distribution of L3 is in part caused by slight ore folding during D4 folding.

D3 deformation has produced significant syn-metamorphic faults - tectonic slides characterized by high strain zones on fold limbs and disruption of folded units on planes sub-parallel to the dominant

foliation S3. This style of deformation is well illustrated in small scale structures found underground. The resultant slicing up of the folded units makes stratigraphic correlations difficult.

- d) D4 Deformation: This produces local, small scale F4 folds which have upright axial planes and plunge 10° - 40° to 100° - 120° . F4 folds fold the S3 foliation and in the more phyllitic rocks develop an axial planar crenulation cleavage. A well developed crenulation lineation L4 plunging 10° - 40° to 100° - 120° is found on S3 surfaces.
- e) D5 Deformation: This is confined to minor reverse kink bands in the hangingwalls of brittle contraction faults. The F5 axial planes generally dip steeply westward striking approximately 340° and the F5 fold axes plunge gently 5° - 10° to 160° - 170° . The D5 faults have gouge on the fault planes indicating high level brittle deformation.

Detailed structural mapping of the 10350E, 10500E and 10670E crosscuts was completed by McClay and Marten and the results are as follows:

- a) 10350E: The ore zone is strongly foliated (S3 foliation) with interleaved foliated marble bands. Fragments of foliated schists occur in the massive sulphides and the hangingwall consists of foliated marble and a black quartzite with foliation parallel sphalerite "veins". Minor F5 kink bands occur in the strongly foliated hangingwall quartzite.

Structural data shows redistribution of poles to S3 about an F4 fold axis plunging 39° - 110° . L3 lineations plunge 44° to 020° and L4 lineations plunge variably from 40° to 090° to 30° to 135° . The massive sulphide ore bands exhibit a "milled" ore texture with elongate vein and schist fragments and are parallel to the S3 foliation.

- b) 10500E: The hangingwall quartzites and quartz-sericite schists are strongly folded by F3 folds with a penetrative S3 foliation. Numerous tectonic slides occur on fold limbs dividing the section into panels of highly folded quartzites and quartz-sericite schists separated by discontinuities-tectonic slides. Exact movement on

the tectonic-slides cannot be determined, in many cases it appears to be reverse movement up-dip to the west.

The quartz-sericite schists are strongly transposed into the S3 foliation. The S3 foliation is very intense and there maybe considerable movement along this foliation surface. In places this movement can be seen by the displacement of quartzite bands. Considerable flattening strain is taken up in the folding and accompanying foliation development.

Many F3 folds are intrafolial, particularly in the limestone and quartzites where S2 fabrics are well displayed. Insufficient S0-S1-S2 relationships were found to allow an interpretation of F2 and F1 fold geometries.

The massive arsenical sulphides are concentrated at the base of the ore zone and sphalerite and quartz are concentrated towards the hangingwall. Fragments of quartz veins and sheared schists are found within the banded arsenopyrite-pyrite portions of the ore zone.

Structural poles to S3 show the foliation dipping from 44° - 60° towards 030° - 060° . L3 lineations are widely distributed, in part reflecting earlier S2 deformation and partly reflecting redistribution by F4 folding. F3 fold axes show a similar distribution whereas L4 (crenulation lineations) plunge from 30° - 40° to 080° - 120° . Late brittle faults (with gouge) have associated upright kink bands.

- c) 10670E: The crosscut passes through intensely folded quartzites and quartz-sericite schists which display well developed F3 folds and S3 foliation. The F3 folds are 1-2 m wavelength comprising irregular folded lenses of quartzites in strongly foliated quartz-sericite schists. The folds are near-isoclinal with a penetrative axial planar S3 fabric. The fold limbs are intensely sheared, resulting in dislocations parallel to S3. S2 spaced foliations are observed in the quartzitic layers. The unusual nature of the fold hinges reflect the effects of earlier (D1 and D2) deformations resulting in boudinaged quartzite layers which are then refolded by F3 folds. The folded quartzite layers have thickened hinge zones indicating

a component of flattening in the D3 deformation. Multiple F3 folds are developed throughout the crosscut, all showing the development of a penetrative S3 foliation. In places, quartz "veins" are transposed into the S3 fabric as are the compositional bands of the quartzite.

Structural data indicates a mean S3 foliation plane dipping 50° to 063° . Minor fold axes are variably developed (on F3 fold limbs), but generally plunge 37° to 117° .

- d) A summary of work done by Selco geologists in the 10820E crosscut is as follows: The S2 banding is the earliest structure observed and is oriented at $108^{\circ}/48^{\circ}$ NE. The S3, which is at $140^{\circ}/48^{\circ}$ is axial planar to the F3 isoclinal folds that fold the S2 banding. Only a few F3 folds were observed, but the low angle between the banding and the foliation suggests isoclinal folding. The folds are best preserved in quartzite bands and plunge 47° to 035° . The S4 foliation, which is axial planar, is oriented at $120^{\circ}/$ vert. A few chevron folds were observed and they fold

both the S2 and S3 and plunge at 20° to 120°. These are F4 folds and generally occur in the phyllitic rocks. The S5 foliation is axial planar and oriented at 108°/60°S. Late brittle kinking represents the final folding phase (F5) and is uncommon.

Since the 830 m drift follows the strike of the Main Zone, the F3 folds are poorly exposed, but the crosscutting F4 folds are well developed. The F4 folds refold the ore zone and fold the dominant S3 foliation with upright axial planes and are generally kink-like except in the more ductile sulphides. There are clear discordances between the ore zone and hangingwall foliations and the footwall foliations. It should be noted that the mean F4 fold axis plot is sympathetic with the plot of the limestone contours (See Plan 22).

McClay summarized by stating that the sheet-like nature of the sulphide body and its structural disposition in the shear zone precludes any major tectonic thickening in the deposit. This type of tectonic regime is characterized by intense flattening

and thinning of units. The thickness variations are rapid and are a result of the heterogenous nature of the shearing strains rather than any geometric structural control. Shearing strains would tend to be concentrated in the ductile sulphides rather than the more competent quartzite and quartz-sericite schists.

Pervasive, sericitic alteration is found within all lithologies and is extensively developed within the Main Zone and its' immediate hangingwall and footwall rocks, but is masked by the footwall limestones. A detailed study of all of the drill sections (crosscuts and J.V.) revealed that this sericitic envelope ranges from 2.1 to 21.3 m wide.

Whether the sericite is due entirely to faulting and shearing or the emplacement of the sulphides is unknown. The most significant observation is that the sericitic halo is found accompanying all of the sulphide horizons at the J & L. The sericite alteration is strong to very strong at the Main and North Zones, but relatively weak at the West and Far East Zones.

Silicification is also found, locally. Limestones, within the Main Zone, are extensively silicified near the end of the 830 drift. There is also a thick siliceous section of quartzites and quartz-chlorite schists above the Main Zone in the 10670E crosscut which maybe, in part, due to silicification.

GEOCHEMISTRY

1. Sampling

A total of 20 heavy mineral concentrate samples were collected from the J & L claim area during the field season. These samples were collected from Kelly Creek, Carnes Creek and McKinnon Creek. The sampling was done in order to determine the distribution of significant elements upstream and downstream of the J & L deposit, to ascertain approximate distances of transport for the various metals so that the information could be applied to reconnaissance sampling in other areas where arsenical-gold deposits might be expected and to try to locate any additional mineralized occurrences within the claim area (See Figure 29).

A total of 56 surface chip samples were collected for assaying from the new showings on Roseberry Mountain (See

Figures 12-28). Three chip samples of the dioritic intrusive were also collected in order to test for low grade precious metal values and was reported from the "Roseberry Prospect". The surface crew chipped 157 lithogeochemical rock samples from L2+00N (82 samples) and L6+00N (75 samples) and collected numerous representative rock samples from Roseberry Mountain.

Sampling from the underground development consisted of samples from the 830 metre level drift, the new crosscuts and the drill core.

Detailed chip sampling of every 2nd round (approximately every 3.5 metres) drift face was completed as the drift advanced. Chip samples of each lithology, with a maximum sample length of one metre, were taken at 3 different horizontal levels across the face. These were at 0.50, 1.30 and 2.10 metres, respectively, above the rail elevation. This sampling was done in order to obtain detailed data on the nature and tenor of the mineralization within the Main Zone. A total of 1248 drift face chip samples were collected (See Appendix II).

Check samples, which consisted of placing 1 tablespoon of material from a previously assayed bulk sample into a sample bag, were put into sample shipments in order to test the reliability of our assay results. A total of 29 check samples were taken. A duplicate chip sample of every 50th drift face sample was also collected and sent to a different assay lab in order to, roughly, test the magnitude of our results. A total of 17 duplicate samples were sent out for assaying. Specific gravity samples were also collected from every 2nd round drift face and these consisted of collecting a representative grab sample of each middle row chip sample.

Lithologic chip samples were also taken, at a height of 1.30 metres above the rail, along the west wall of the 2 crosscuts. Detailed, three level, chip sampling was also done in the 10820E crosscut, where significant sulphide mineralization was encountered. A total of 102 chip samples from the 10670E crosscut and 158 chip samples from the 10820E crosscut, were collected.

Other samples collected included 2 grab samples from the drift and 16 detailed chip samples of 2 selected, folded

massive sulphide bands. The grab samples included one massive sulphide sample which was pyrrhotite-rich and one of very coarse-grained arsenopyrite. The pyrrhotite samples was taken to test for minor accessory elements and the arsenopyrite sample was collected in order to test the relationship of precious metal content to arsenopyrite grain size. The fold samples were taken in order to investigate the distribution of assayed elements with respect to fold noses and limbs (See Figures 30 and 31).

Sampling of the underground workings, during 1984, amounted to 1543 samples. Numerous, representative grab samples were collected from the drift and the two crosscuts. One 5-gallon pail was also filled with massive sulphides from the muck pile of every second round blast in the 830 metre drift. These were collected and stored in order to have samples available for possible future metallurgical testing.

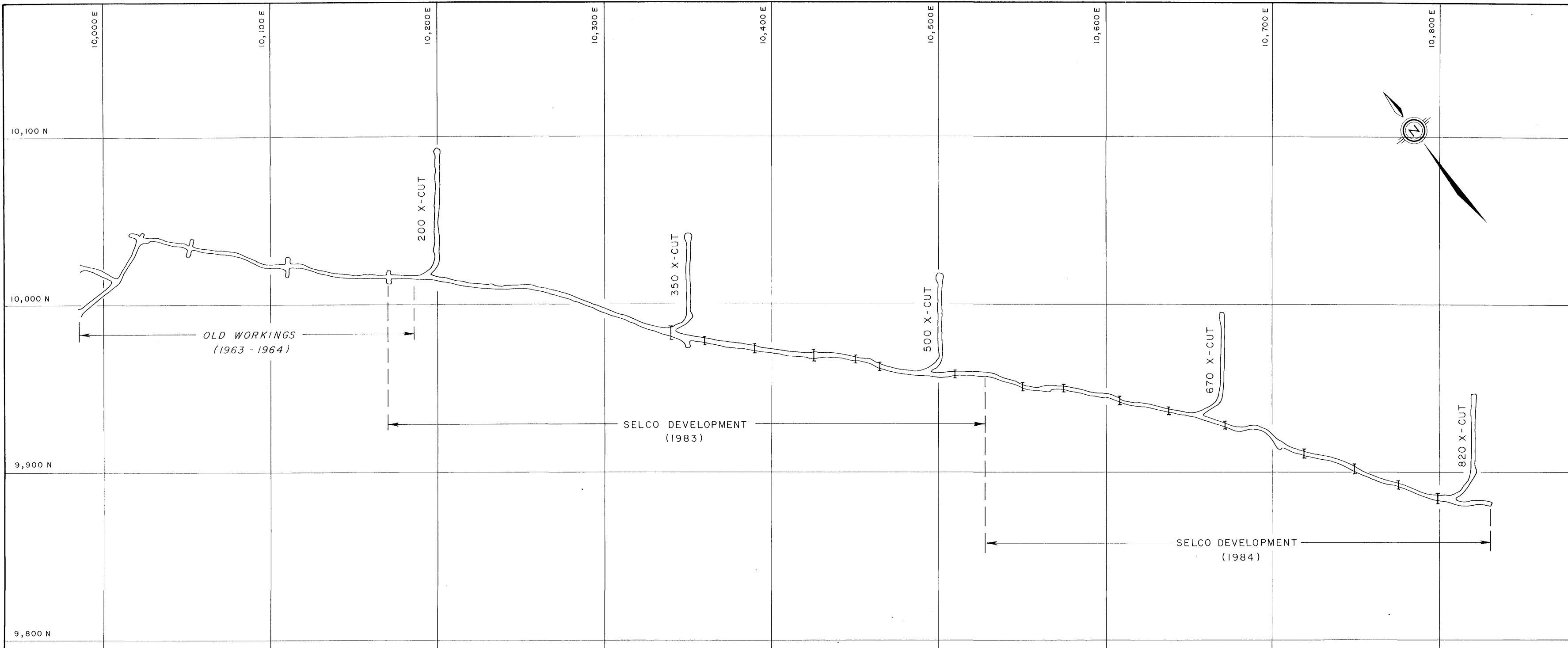
A metallurgical sample, which is being used this year, was collected at 16 strategic locations from within the 1983 and 1984 underground development. These samples consisted of continuous chip samples across the drift back and each

sample filled, approximately, half of a 5-gallon sample pail. The samples were taken from the following locations: 10340E, 10360E, 10390E, 10425E, 10450E, 10465E, 10510E, 10550.6E, 10574.505E, 10607.92E, 10637.73E, 10671.65E, 10718.442E, 10749.64E, 10774.81E, 10799.48E. These samples should be reasonably representative of the Main Zone mineralization at the 830 metre level (See Figure 9).

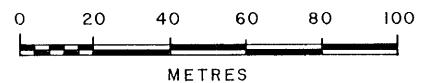
Sampling of the drill core consisted, primarily, of split/cut massive sulphide sections, the intervening country rock and at least one metre of each of the hangingwall and footwall lithologies, on either side of the mineralization. These core samples were taken according to lithology and a maximum sample length of one metre was taken. A total of 291 core samples were collected.

2. Analysis

In an attempt to reduce costs, only the chip samples from the new surface showings and the intrusive samples were sent to the lab. Underground, only the middle row of chip samples from unmineralized drift faces and both the middle and top rows of chip samples from the 10820E crosscut mineralized section and the mineralized drift faces were



I SAMPLE SECTIONS
(taken across drift back)



SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT METALLURGICAL SAMPLE LOCATIONS 830m LEVEL DRIFT	
SCALE 1:2,000	DRAWN BY: R. PEGG FIG. 9
DATE JANUARY 1985	DRAFTED BY: L.G.
N.T.S. 82 M / 8E	PROJ. 10120 REPORT BPVR 84-53

sent out for assaying. Thus, only 1343 samples were assayed/geochemed and this included 59 surface chip samples, 877 drift face samples, 52 crosscut samples, 16 fold nose samples, 29 check samples, 17 duplicate samples, 2 representative samples and 291 drill core samples. The remaining samples and drill core are being stored in Castlegar, BC.

The samples were shipped from Revelstoke to Chemex Labs Limited of North Vancouver for analysis. Most of the samples were assayed for Pb, Zn, As, Ag and Au. The pyrrhotite-rich representative sample was geochemically assayed for Sn, W, Sb, Cd, Ni and Bi.

The duplicate samples were shipped from Revelstoke to Min-En Labs Ltd. of North Vancouver and were assayed for Pb, Zn, As, Ag and Au.

The Chemex assay methods are as follows:

- a) Pb-Zn(%): A 2 gram sub-sample is digested in hot perchloric-nitric acid mixture for two hours, cooled, then transferred into a 250 ml volumetric flask.

Nitric acid is added to the final sample and standard solutions. The solutions are then analyzed on an atomic absorption instrument.

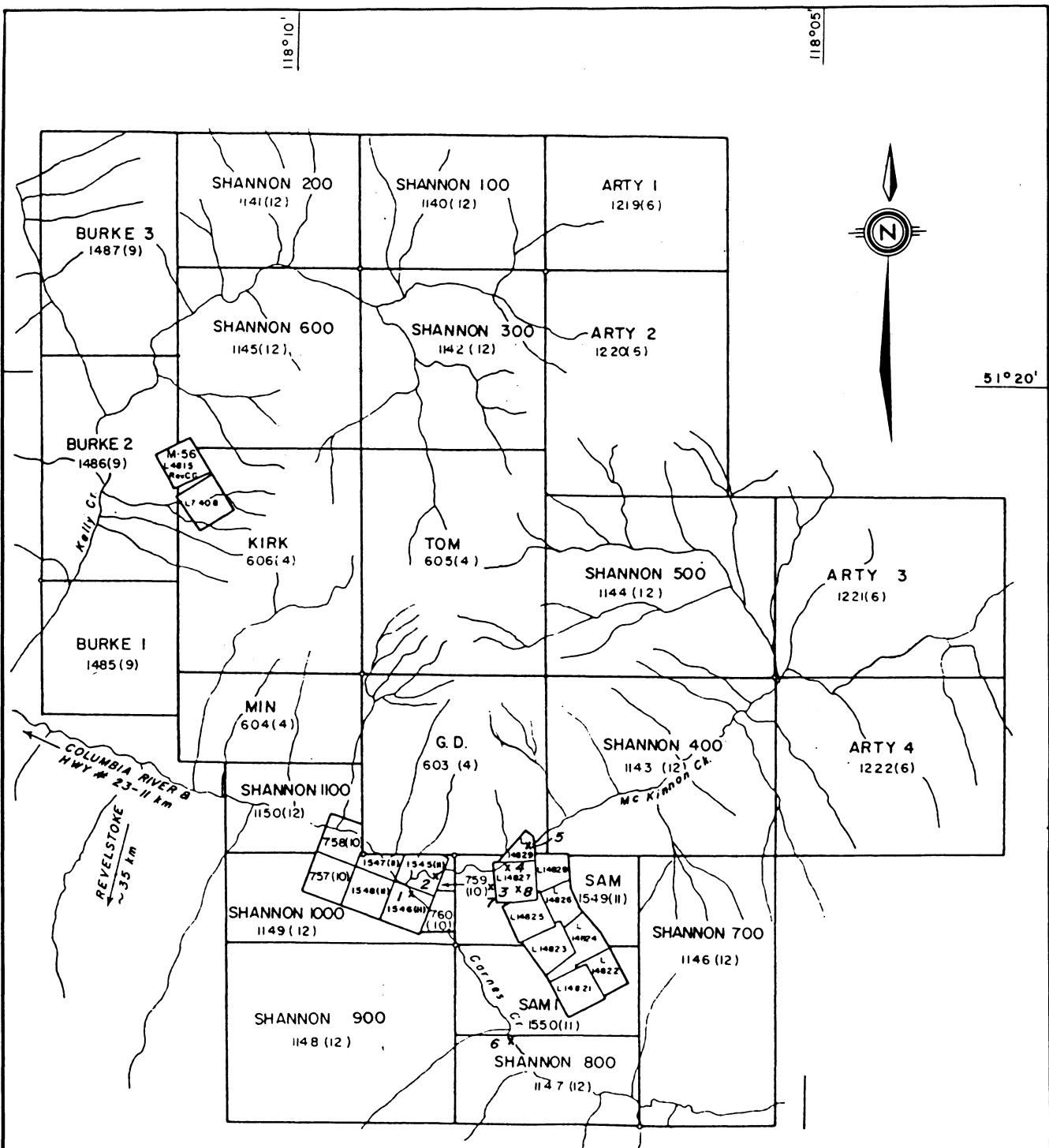
- b) Ag-Au(g/t): Silver and gold analyses are done by standard fire assay techniques. In the sample preparation stage the screens are checked for metallics which, if present, are assayed separately and calculated into the results obtained from the pulp assay. The 0.5 assay ton sub-samples are fused in litharge, carbonate and siliceous fluxes. The lead button containing the precious metals is cupelled in a muffle furnace. The combined Ag and Au is weighed on a microbalance, parted annealed and again weighed as Au. The difference in the two weighings is Ag.
- c) Arsenic(%): A one gram sample is irradiated in a thermal neutron flux. The gamma activity of the resulting arsenic isotopes is determined by gamma spectroscopy to quantify the arsenic content to the detection limit of 0.001%

3. Environmental Monitoring

Water samples were again collected twice a month from both Carnes and McKinnon Creeks, both up and downstream from the area influenced by the underground exploration program (See Figure 10). One sample of rain water and one from the water draining diamond drill hole 83-12 were collected to obtain background results. A total of 69 samples were collected this year.

Samples were collected in plastic containers and shipped to Barringer Magenta Ltd. of Calgary, for analysis. All samples were analyzed for Cu, Pb, Zn, Fe, As, Hg, pH, Alkalinity and Hardness. The second, monthly sample was additionally analyzed for dissolved solids, Br⁻, Cl⁻, F⁻, NO₂⁻, NO₃⁻; PO₄³⁻ and SO₄²⁻. This analysis was recommended by the Ministry of the Environment. Notes were also kept on the weather and creek conditions which could affect results.

Results to date indicate a normal annual fluctuation of element levels within drainage waters at the J & L. Highest values for "contaminants" is in the spring months of March, April and May during runoff conditions. The



LEGEND

- x 1, 6 ... Carnes Ck. Water Sample Location
- x 2, 5 ... McKinnon Ck. Water Sample Location
- x 3 ... Drinking Water Sample Location (Spring)
- x 4 ... Adit Water Sample Location
(Location of U/G Activity)
- x 7 ... Rain Water
- x 8 ... D.D.H. 83-12

0 1 2 3
KILOMETRES

BP SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT WATER SAMPLE LOCATIONS	
SCALE AS SHOWN	DRAWN BY: R. PEGG
DATE NOV. 1984	DRAFTED BY: J.S.
N.T.S. 82 M / 8E	PROJ. 10120
REPORT BPVR 84-53	

August 16, 1983 sample (Site 4) reflects increased site development activity underground with track rehabilitation with a resultant increase in contaminants to the mine discharge water. Sites 1 and 2 identify abnormal runoff conditions during mid-October 1983 with an associated increase in total dissolved solids.

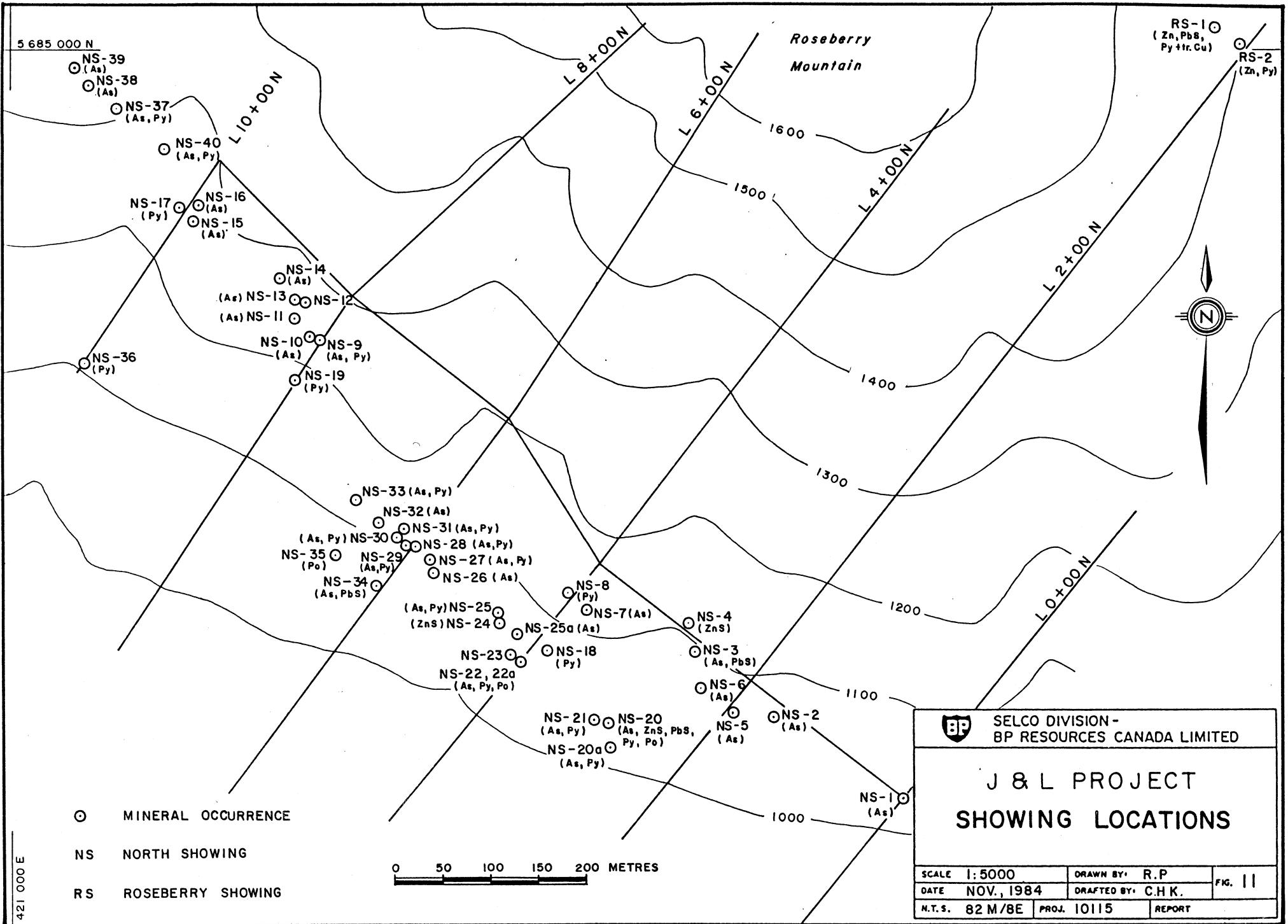
In conclusion, no significant contamination of the drainages has been identified due to active exploration work at the J & L. Levels of all "poisonous" substances such as arsenic, mercury, lead, etc, show only natural fluctuations with no abnormal concentrations or increasing trends with time.

Water sampling of the all drainage sites will continue during 1985 as field conditions permit.

ECONOMIC GEOLOGY

1. Surface Showings - Roseberry Mountain

During 1984, the surface crew located 44 mineral occurrences on the south slopes of Roseberry Mountain (See Figure 11) and extended the North Zone for 1150 metres to



the northwest of the original North Zone showing (NS-1). Forty of the occurrences, which are comprised essentially of arsenopyrite and/or pyrite with varying amounts of scorodite and jarosite, are hosted by quartzites and sericitic phyllites of the Hamill Group. Hydrozincite, sphalerite, galena, chalcopyrite and pyrrhotite were also observed, locally, in small amounts. The host rocks are well sheared and exhibit a strong to moderate, narrow sericite halo which was also observed along the main zone on Goat Mountain. It should also be noted that a discontinuous, grey banded limestone bed was observed in the immediate footwall at NS-20 and NS-24.

The above showings were broken down into 4 groups (See Table III). The Upper and Lower Showings are the most significant and they trend at $148^{\circ}/50^{\circ}$ NE, subparallel to one another and are, approximately, 85 metres apart, horizontally. These showings consist of between 1 and 4 massive arsenical bands and/or stringer zones which vary between 0.005 and 1.270 metres, estimated true width. The arsenical massive sulphide bands are oxidized and leached to varying degrees, narrow and discontinuous and appear, locally, to grade into arsenical stringers and/or gossaneous pyritic zones.

TABLE III

1984 SURFACE SHOWING DESCRIPTION - ROSEBERRY MOUNTAIN

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph. zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
Hangingwall Showings: NS-1 to NS-4, inclusive (hosted by Hamill Group stratigraphy)										
NS-1	x (1983)	1030*	0.050	10.2	Pinched out in both directions.	150°	Massive As plus scorodite in a quartz gangue.	Moderately oxidized	Sheared dirty quartzite	Sheared dirty quartzite
NS-2	x	1080 (1070)*	0.070-0.040	4.0	Covered by overburden in both directions.	150°	Top 2 cm - Massive sulphide - 40% As plus scorodite in a quartz-sericite gangue with trace Py. Lower 5 cm - 10-20% disseminated As plus scorodite in a quartz-sericite gangue.	Moderately to heavily oxidized (scorodite)	Sericite-Quartz phyllite - highly fissile, highly friable - iron oxides on foliation planes - bleached	Sericite-Quartz phyllite - highly fissile, highly friable - iron oxides on foliation planes - bleached
NS-3	x	1090 (1095)*	0.090-0.000	7.0	Pinches out to the west, covered by overburden to the east.	140°	Massive sulphide - 80% As, 5% Py, <5% PbS, plus 0-10% scorodite in a quartz-sericite gangue.	Fresh, except locally up to 10% scorodite.	Quartz-Sericite phyllite - moderately fissile - minor disseminated Py plus very small Py stringers	Sericite phyllite - minor interbanded quartz-sericite phyllite - moderately fissile
NS-4		1115*	0.100	approximately 20.0	Pinches out in both directions.	140°	Narrow leached ZnS stringers plus hydrozincite with a speck of PbS - minor Py plus Po occurs in quartz lenses.	Oxidized, calcareous (hydro-zincite)	Quartz-Sericite phyllite - moderately to highly fissile	Quartz-Sericite phyllite - moderately to highly fissile
Upper Showings: NS-5 to NS-17 and NS-37 to NS-40, inclusive (hosted by Hamill Group stratigraphy)										
NS-5		1065*	0.020-0.010	zone may continue for 30 m to NS-6	Covered by overburden to the east, and could not be traced to the west due to a steep cliff.	128°	Massive sulphide - 30% fine-grained to coarse grained As plus 65% scorodite in a quartz-sericite gangue, zone gets pinched out by barren quartz vein lenses.	Heavily oxidized (scorodite)	Sericite phyllite - minor quartz component - heavily iron stained	Calcareous grit - thin sericitic bands in a dark reddish-brown banded unit.
NS-6		1065*	0.010-0.000	3.0, but the horizon may be cont. to	Pinched out in both directions.	130°	Massive sulphide - 30% fine grained to coarse-grained As plus 65% scorodite in a quartz-sericite gangue.	Heavily oxidized (scorodite)	Sheared phyllite - moderately iron stained	Sheared sericite phyllite - moderately iron stained

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph.zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
NS-7		1125*	0.015-0.005	7.5	Pinched out to the east, covered by overburden to the west.	146°	Massive As plus minor scorodite in a quartz gangue	Weakly oxidized (scorodite)	Sheared sericite phyllite - heavily iron stained - minor quartz veins with minor chlorite	Sheared sericite phyllite - heavily iron stained - clay alteration
NS-8		1130*	N.A.	N.A.	N.A.	N.A.	Py stringers approximately 1%	Heavy jarosite plus limonite staining	Clean Quartzite - heavily iron stained - minor quartz veins with minor chlorite	Clean Quartzite - heavily iron stained - minor quartz veins with minor chlorite
NS-9	x	1270 (1240)*	0.100-0.050	10.0	Pinched out to the west, covered by overburden to the east.	140°	Massive As with minor scorodite, jarosite plus limonite. Minor intercalated jarosite stained, bleached quartzite with trace Py.	Minor oxidation (scorodite, jarosite, limonite)	Quartzite - highly bleached - clean - iron stained on weathered and fresh surfaces	Quartzite - foliated - moderately fissile - sericite on foliations - locally bleached - 3% limonite
NS-10		1240*	0.030	0.25	Pinched out in both directions.	144°	Massive As with minor scorodite in a quartz gangue.	Minor oxidation (scorodite)	Quartzite - heavily iron stained	Quartz-Sericite phyllite - heavily iron stained
NS-11	x	1270 (1245)	0.050-0.005	12.5	Pinched out to the west, covered by overburden to the east.	148°	Massive sulphide - 15-30% As plus 50-70% scorodite in a quartz gangue.	Heavily oxidized (scorodite)	Sericite-Quartz phyllite - moderately fissile - minor limonite horizons	Sericite phyllite - moderately to highly fissile - 15% boudinaged quartz veins - minor quartzite

Note: The following three showings (NS-12, 13, 14) are all part of the same horizon that has a total observable strike length of 59 m

NS-12		1265*	0.100-0.000	4.0	Pinched out in both directions.	130°	Massive As with up to 70% scorodite in a quartz gangue.	Heavily oxidized (scorodite)	Quartzite - bleached, sheared - jarosite stained	Quartzite - bleached - jarosite stained
NS-13	x	1280 (1265)*	0.220-0.005	9.0	Covered by overburden in both directions.	130°	Massive sulphide - 15-50% As, 40-80% scorodite in a quartz gangue. The zone grades laterally into As stringers within highly bleached, sheared quartzite.	Heavily oxidized (scorodite)	Quartzite - clean, bleached - moderate limonite plus jarosite staining - trace Py	Quartzite - bleached - minor sericite phyllite - locally up to 5% limonite
NS-14	x	1280 (1275)*	0.080-0.020	4.1	Covered by overburden in both directions.	144°	Massive sulphide - 70% As plus 15% Py in a quartz gangue.	Fresh	Quartzite - dirty with minor sericite phyllite - cleaner quartzite near mineralized horizon - 2% limonite after Py - trace Py	Quartzite - clean, massive - 1-2% coarse grained Py - weakly iron stained
NS-15		1265*	0.070-0.020	1.2	Pinched out in both directions.	140°	Massive sulphide - 30% As plus 50% scorodite in a quartz gangue.	Heavily oxidized (scorodite)	Sericite - Quartz phyllite - moderately fissile - moderately iron stained	Sericite - Quartz phyllite - moderately fissile

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulphide zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
NS-16		1275*	<0.010	<1.0	Pinched out to the west, covered by overburden to the east.	138°	Massive As with moderate scorodite	Moderately oxidized (scorodite)	Quartz - sericite phyllite - minor quartzite	Quartz - sericite phyllite - minor quartzite
NS-17		1260*	<0.010	N.A.	N.A.	135°	Two gossanous pyritic horizons with approximately 2% disseminated Py in blueish-white ultra clean quartzite.	Heavy iron staining (limonite, hematite, jarosite)	Quartzite - clean, bleached - minor bands of sericite- quartz phyllite - iron stained	Sericite - Quartz phyllite - minor interbanded clean quartzite
NS-37 (same horizon as NS-38)	x	1350 (1315)*	Zone 1: 0.250 (at fold nose)-0.005 (on fold limbs) Zone 2: 0.180-0.140	65.0	Covered by overburden in both directions.	116°	Zone 1: Massive sulphide - 30% medium grained As plus 20% coarse grained Py in a quartz gangue. Zone 2: Massive sulphide - 85% As in a quartz gangue.	Moderately weathered (scorodite) Fresh (minor scorodite)	Quartzite - clean, bleached - locally contains minor As plus Py as thin stringers and disseminated grains	Quartzite - clean, bleached - minor sericite on foliation - contains 1-2 cm wide massive As horizon
NS-38 (same horizon as NS-37)	x	1370 (1335)*	0.200-0.140	65.0	Covered by overburden in both directions.	116°	Two bands of massive sulphide, one on top of the other. The lower band contains 70% As in a quartz gangue. The upper band contains 50% As in a quartz gangue with minor interbanded bleached quartzite.	Upper zone is moderately oxidized (scorodite) Lower zone is fresh	Quartzite - clean, massive, bleached - minor As stringers	Quartzite - cleaned, bleached - minor sericite on foliation planes - minor jarosite, plus limonite staining
NS-39		1360 (1355)*	0.015-0.002	3.8	Covered by overburden in both directions.	106°	Thin band of massive As	Weakly oxidized (scorodite)	Quartzite - very clean - hematite stained - minor As stringers	Quartzite - clean, highly bleached - minor sericite on foliation planes
NS-40		1270*	0.020-0.010	10.0	Covered by overburden to the west, could not be traced east due to steep cliff.	140°	Two thin bands of massive As with minor Py.	Weakly oxidized (iron oxides)	Quartzite - clean - minor interbanded quartz- sericite phyllites	Quartzite - clean - minor interbanded quartz-sericite phyllite - heavily iron stained
Middle Showings: NS-18 and NS-19, inclusive (hosted by Hamill Group stratigraphy)										
NS-18		1075*	<0.010	N.A.	N.A.	142°	Gossanous pyritic zone with approximately 3% Py in stringers.	Iron stained	Quartzite - clean, massive - brown	Quartzite - clean, massive - brown

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph. zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
NS-19		1185*	<0.010	N.A.	N.A.	127°	Gossaneous pyritic zone with approximately 8% Py in stringers.	Heavy iron staining (jarosite, hematite, scorodite?)	Quartzite - clean, massive	Quartzite - clean, massive
Lower Showings: NS-20a, NS-20 to NS-22, NS-22a, NS-23 to NS-24, NS-25a, NS-25 to NS-33, inclusive (hosted by Hamill Group stratigraphy)										
NS-20a (old workings)		1020 (1014)*	0.400	6.4	Covered by overburden in both directions.	113°	As, plus Py stringers in ultra clean quartzite with minor sericite on foliation planes - 2% As, 2% Py	Fresh	Quartzite - pervasively iron stain - massive to semi massive	Sericite phyllite - heavily stained with limonite plus minor jarosite - bleached
NS-20 (same horizon as NS-21)	x	1020 (1035)*	Zone 1: 0.220* Zone 2: 0.160* Zone 3: 0.310* Zone 4: 0.660*	<4.0 <4.0 69.0 <4.0	Most of the zones pinch out in both directions. Zone 3 is pinched out to the west and covered by overburden to the east.	138°	Zone 1: Disseminated fine grained PbS and ZnS associated with boudined quartz veins Zone 2: Massive arsenopyrite with minor scorodite, Py, jarosite and chlorite. Zone 3 & 4: Bands of massive As (up to 4 cm wide), hosted in quartzite and sericite phyllite respectively, with trace PbS, ZnS, Py and scorodite. Along strike, a band from Zone 3 locally contains Po (up to 50%) and trace chalcopyrite.	Fresh Moderately oxidized (scorodite) Fresh	Quartz - sericite phyllite - bleached, sheared - talc plus clay alteration in sheared zones - heavy limonite plus jarosite staining	Sericite - phyllite - carbonate fracture filling - underlain by limestone Limestone - grey, thinly banded - minor limonite staining - pinches down to 20 cm below NS-21 and pinches out completely 2 m west of NS-21 - does not outcrop to the east
NS-21 (same horizon as NS-20)	x	1040 (1035)*	0.810*	69.0	Pinched out to the west covered by overburden to the east.	138°	Highly sheared, bleached sericite phyllite with 5% As - 3% massive As stringers, 2% disseminated As with trace ZnS and PbS and 10% scorodite in massive zones. Zone slumped 22 cm down.	Heavily oxidized (jarosite, limonite)	Sericite phyllite - minor interbanded quartzite - weakly iron stained - 10% boudined quartz veins - moderately fissile - locally contains minor talc	Quartzite - locally bleached - moderate jarosite staining - moderately fissile with sericite on foliation planes - contains trace As stringers - trace disseminated Py
NS-22	x	1110 (1050)*	Zone 1: 1.270* Zone 2: 0.900*	13.5	Covered by overburden to the east and appears to be terminated by a fold nose to the west.	140°	Zone 1: Thin As and Py stringers plus disseminated As and Py in clean, white massive quartzite. Zone 2: Trace As and Py as stringers and disseminated grains in bleached, weakly foliated quartzite.	Fresh Fresh	Quartzite - bleached, clean - trace disseminated Py - limonite on fractures	Quartz-sericite phyllite - moderately fissile - trace As stringers
NS-22a		1050*	Zone 1: 0.005-0.000 Zone 2: 0.010-0.050 Zones are 8 m apart	2.0 5.5	Both zones are pinched out in both directions.	140°	Zone 1: Massive sulphide 80% As, 20% Py plus minor Po. Zone 2: Massive As with trace Py	Fresh Fresh	Quartzite plus sericite phyllite - heavily iron stained	Quartzite plus sericite phyllite - heavily iron stained

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph. zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
NS-23		1050*	0.040	1.8	Pinched out to the east, covered by overburden to the west.	150°	Massive sulphide - 10% As, 85% scorodite that is locally disrupted by quartz veins.	Highly oxidized (scorodite)	Quartz-sericite phyllite	Quartz-sericite phyllite - iron stained
NS-24	x	1070 (1060)*	0.200	12.0	Covered by overburden in both directions.	145°	Bands of red ZnS (approximately 7%) within fine grained, weakly banded limestone.	Fresh	Sericite phyllite - minor quartz component - highly fissile	Limestone - fine to medium grained with coarse calcite veins - minor disseminated, coarse grained Py - minor hydrozincite
NS-25		1065*	0.025-0.010	10.0	Pinched out in both directions. Small arsenical 38 m east along strike.	146°	Massive sulphide - 80% coarse and fine grained As within a quartz gangue. The unit splits along strike into several small (<1 cm) horizons and locally grades along strike into As within highly gossanous, bleached quartzite.	Fresh	Quartzite - highly foliated - sericite on foliations - oxidized	Quartzite - clean - oxidized, brown
NS-25a		1070*	0.110-0.010	36.3	Pinched out to the west, covered by overburden to the east.	148°	Several coalescing bands of massive sulphide - up to 80% coarse grained As in a quartz gangue. Bands pinch and swell along strike.	Fresh to highly oxidized (scorodite)	Sericite - Quartz phyllite - oxidized - moderately fissile	Sericite phyllite - moderately oxidized - highly fissile
NS-26	x	1096 (1095)*	0.14*	4.4	Covered by overburden in both directions.	146°	Massive sulphide - 40% fine to coarse grained As plus 40% scorodite in a quartz gangue.	Moderately oxidized (scorodite)	Quartzite - massive, clean - bleached - minor Py in stringers and as disseminated grains	Quartz-Sericite phyllite - highly fissile - minor limonitic horizons plus crosscutting limonite veins - trace Py
NS-27		1105*	0.040-0.005	17.0	Pinched out to the west, covered by overburden to the east.	158°	Massive sulphide - 70% coarse and fine grained As plus 20% coarse grained Py. In one locality the massive As is deflected around a large (30 cm wide, 1 m long) boudinaged quartz vein.	Fresh except moderately oxidized around the quartz vein (scorodite).	Sericite-Quartz phyllite - jarosite staining - bleached - highly fissile	Quartzite - bleached - jarosite staining
NS-28	x	1112 (1120)*	0.190-0.010	22.0	Pinched out to the west, covered by overburden to the east.	122°	Massive sulphide - 60% fine to medium grained As plus 30% scorodite in a quartz gangue. Horizon pinches and swells along strike.	Moderately oxidized (scorodite)	Quartz-Sericite phyllite - highly fissile - iron oxides on crosscutting fractures.	Quartz-Sericite phyllite - minor limonite staining

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph. zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
NS-29		1120*	0.030-0.010	0.95	Pinched out in both directions.	120°	As stringers (approximately 1 mm wide) locally associated with quartz veins.	Fresh	Quartz-Sericite phyllite - moderately fissile	Quartz-Sericite phyllite - moderately fissile
NS-30	x	1125*	Zone 1: 0.170-0.030 Zone 2: 0.150-0.030	4.3	Pinched out to the west, cut off to the east by a fold nose?	115°	Zone 1: Massive sulphide - 95% As with minor scorodite plus quartz - zone swells greatly at a fault. Zone 2: Massive sulphide - 15% As within a band of 80% scorodite within a quartz gangue that contains minor sericite.	Weakly oxidized (scorodite) Heavily oxidized (scorodite)	Sericite-Quartz phyllite - highly fissile - overlain by massive, clean locally bleached quartzite - trace disseminated As	Quartz-Sericite phyllite - moderately to highly fissile - locally siliceous
NS-31		1130*	0.030-0.010	17.0	Pinched out in both directions.	131°	Massive As plus scorodite	Moderately oxidized (scorodite)	Sericite-Quartz phyllite	Sericite-Quartz phyllite
NS-32		1125*	0.110-0.005	35.0	Pinched out in both directions.	132°	Massive As plus scorodite	Moderately oxidized (scorodite)	Quartz-Sericite phyllite	Quartz-Sericite phyllite
NS-33		1135*	1.200-0.100	7.5	Covered by overburden in both directions.	152°	Sheared zone with 2% As in stringers and pods with minor Py + scorodite. Heavy jarosite staining plus an unknown black stain (Mg?). Hosted in sheared bleached quartzites.	Moderately oxidized (scorodite, jarosite)	Quartzite - clean - heavily iron stained	Quartzite - clean - moderately iron stained - less sheared than mineralized horizon
Intrusive Showings: NS-34 to NS-36, inclusive										
NS-34		1080*	0.020	N.A.	N.A.	N.A.	coarse grained PbS crystals in the centre of coarse grained, eye shaped calcite lenses (10 cm wide, 30 cm long). Trace As occurs as medium grained crystals associated with PbS.	Fresh	Diorite - chlorite alteration	Diorite - chlorite alteration
NS-35		1080*	0.100	2.0	Pinched out to the south continues up cliff to the north.	020°	Massive Po associated with a quartz vein.	Fresh with hematite on weathered surfaces	Diorite - chlorite alteration - heavily iron stained	Diorite - chlorite alteration - heavily iron stained
NS-36		1140 (1115)*	0.60	3.0	Pinched out to the east.. Continues west over a cliff.	127°	Gossaneous zone with quartz veins and 3% Py in thin stringers.	Heavily oxidized (hematite, limonite)	Diorite - dark (black) - iron stained	Diorite - dark (black) - iron stained

Showing No.	Mapped and Sampled	(m) Elev. (*Elev. from map)	(m) Maximum-Minimum true thickness of sulph. zone (*true thickness at sample face)	Observed Strike Length (m)	Terminations	Surface Trend of Mineralization	Description of Sulphide Zone	Condition of Sulphide Zone	Description of Hangingwall	Description of Footwall
ROSEBERRY SHOWINGS: RS-1 and RS-2, inclusive (hosted within the Badshot Formation)										
RS-1	x two sections were mapped and sampled.	1430*	3.000-0.020	65.0	Pinched out in both directions.	106°	Disseminated coarse grained ZnS (1-5%) (honey-orange), PbS (trace) and Py (<1%) with trace amounts of tetrahedrite. The mineralization is concentrated within sheared sericite phyllite and within limestone proximal to the sheared phyllite.	Fresh (minor hydrozincite)	Limestone - grey banded - white - minor thin quartz carbonate veins - contains minor sericite bands on foliation planes	Limestone - grey banded - fine-coarse grained - vaguely banded - minor thin sericite bands - minor crosscutting quartz-carbonate veins and local massive white quartz lenses
RS-2		1400*	0.100-0.010	12.0	Pinched out to the east, covered by overburden to the west.	110°	Massive sulphide - 50% dark red ZnS with 4% PbS plus minor Py in a quartz carbonate gangue within a light green sheared sericite phyllite. The horizon grades along strike into thin ZnS plus PbS stringers.	Fresh (minor hydrozincite)	Limestone - grey, massive - vaguely banded - medium to coarse grained - minor thin crosscutting carbonate veinlets.	Limestone - moderately fissile - vaguely banded, grey - interbedded white limestone plus sericite phyllite

Fifteen of the more significant mineral occurrences were mapped in detail and chip sampled (See Figures 12-28). Results indicate an appreciable gold content over narrow widths and in a rough correlation between a higher base metal content and the presence of the limestone footwall (Lower Showings). The most significant results were obtained from NS-11 (0.01% Pb, <0.01% Zn, 1.306% As, 0.7 g/t Ag and 1.2 g/t Au over 0.82 metres (t.w.)), NS-20 (0.27% Pb, 0.09% Zn, 1.881% As, 20.8 g/t Ag and 3.1 g/t Au over 2.11 metres (t.w.)), NS-37 (0.01% Pb, <0.01% Zn, 4.559% As, 5.4 g/t Ag and 3.0 g/t Au over 1.37 metres (t.w.)) and NS-38 (0.08% Pb, <0.01% Zn, 5.590% As, 10.8 g/t Ag and 1.1 g/t Au over 0.93 metres (t.w.)). It should be noted that NS-20 is the site of an old trench and is probably the mineral occurrence on Roseberry Mountain which was noted in the old J & L reports.

It appears that the Lower Showings have been cut off by the thrust fault and/or dioritic intrusive and that the North Zone is the strike extension of the J & L Main Zone which is found on Goat Mountain (See Plan 3).

The three showings within the diorite are related to calcite lenses and/or quartz veins while the 2 mineral

LEGEND

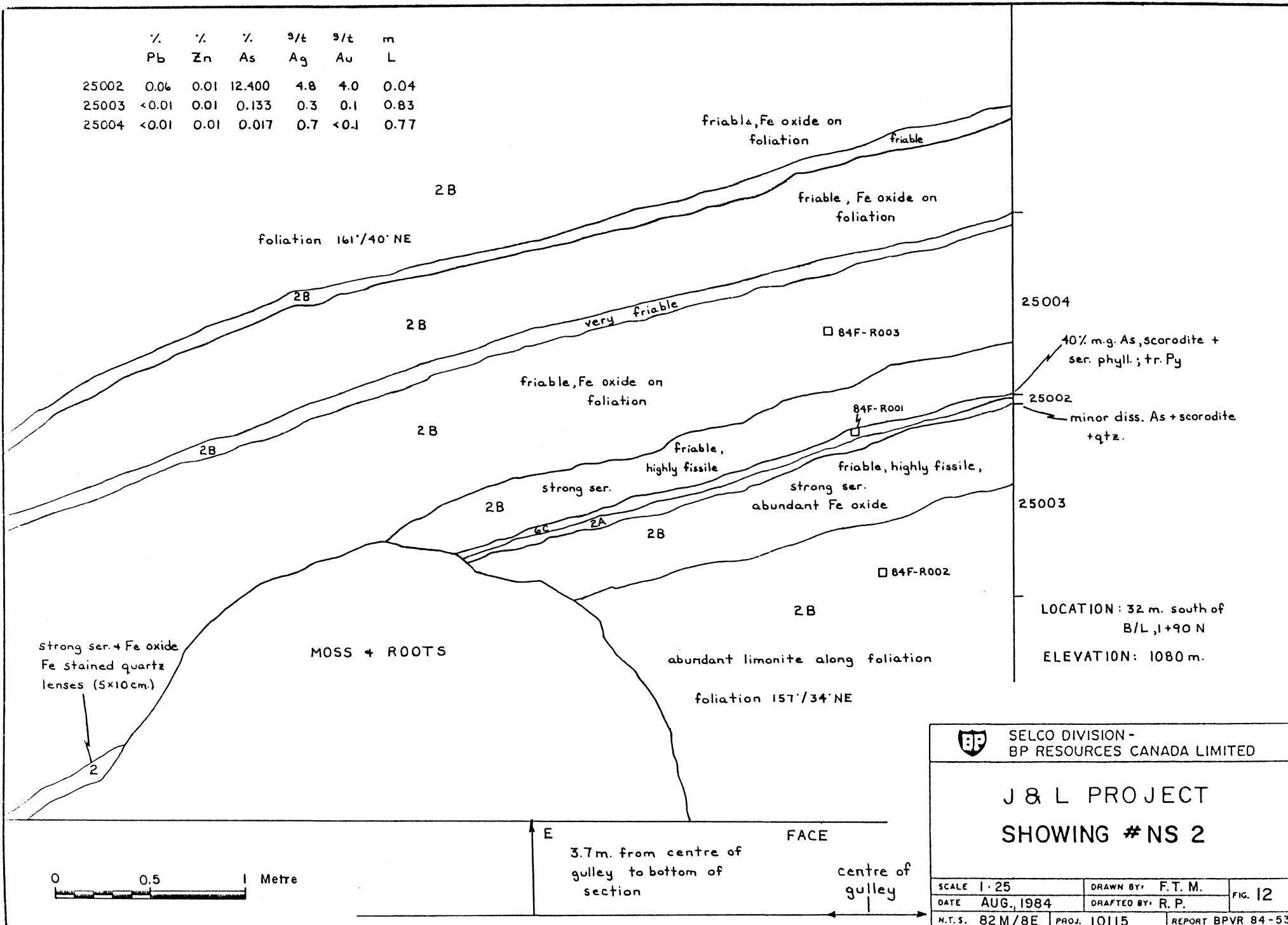
- 1 Blue-grey, crystalline banded limestone, barren
- 1A Limestone - argillite, dark blue-grey, graphitic, barren
- 1B Silicified Limestone, dark blue-grey, cherty, irregular banding
- 2 Sericite schist, buff-white, finely foliated, commonly iron-stained
- 2A Sericite with disseminated sulphides, commonly in a blue-grey siliceous matrix
- 2B Sericite-quartz phyllite
- 3 Quartz - sericite schist, buff to light green, quartz is crystalline, foliation commonly coarse
- 3A Quartz with disseminated sulphides, commonly in a blue-grey siliceous matrix
- 4 Chlorite schist, dark green, finely foliated, minor quartz
- 4A Quartz - chlorite schist, dark-light green depending on chl : ser. content, foliation relatively fine
- 5 Quartzite, buff to grey crystalline quartz \pm feldspar, sericitic foliation planes
- 5A Quartzite, chloritic partings, \pm sericitic \pm feldspar, buff to grey
- 6 Sulphide zone (arsenopyrite + pyrite)
- 6A Massive sulphides in blue-grey siliceous matrix, generally As + Py
- 6B Lenses and bands of sulphides in white quartz
- 6C Partially oxidized sulphides, As + Py + scorodite generally present with quartz \pm sericite \pm talc
- 6D Oxidized sulphides, usually soil from residual weathering, mainly iron oxides \pm scorodite \pm graphite
- 7 Chert colour variable, light grey dark blue. Generally well fractured

82B - R054 Representative Rock Sample Location

|←8613→| Chip Sample Location

m.L Estimated true width

	%	%	%	g/t	g/t	m
	Pb	Zn	As	Ag	Au	L
25002	0.06	0.01	12.400	4.8	4.0	0.04
25003	<0.01	0.01	0.133	0.3	0.1	0.83
25004	<0.01	0.01	0.017	0.7	<0.1	0.77



SELCO DIVISION -
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J & L PROJECT SHOWING # NS 2

SCALE 1:25

DATE AUG., 1984

N.T.S. 82 M/8E

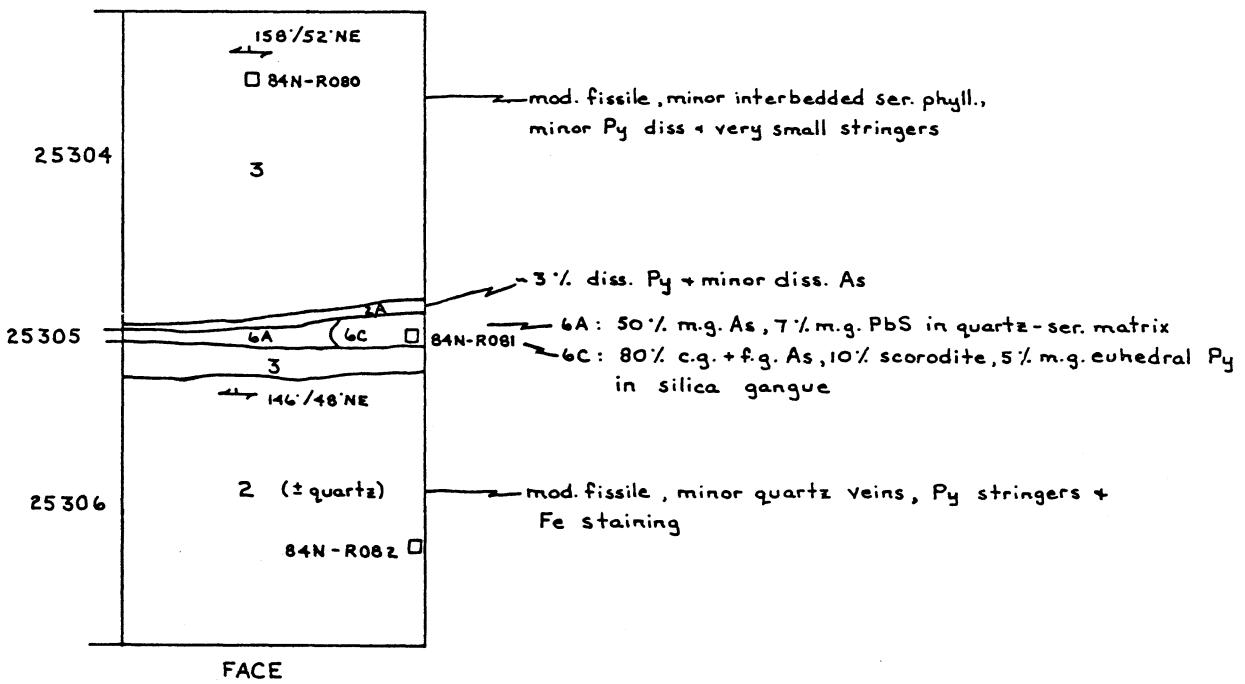
DRAWN BY: F.T.M.

DRAFTED BY: R.P.

PROJ. 10115

FIG. 12

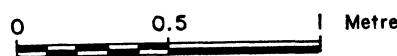
REPORT BPVR 84-53



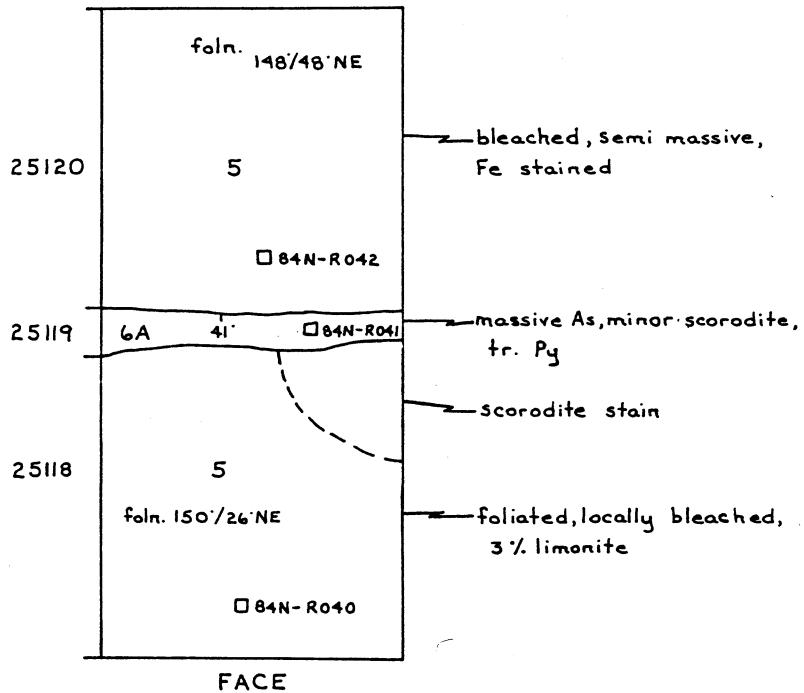
	%	%	%	g/t	g/t	m.
	Pb	Zn	As	Ag	Au	L
25304	0.01	0.01	0.039	1.7	0.1	0.70
25305	1.60	0.01	18.400	33.0	4.0	0.09
25306	0.04	0.01	0.232	0.3	<0.1	0.73

LOCATION: 19 m. west of B/L, 2+50N

ELEVATION: 1090 m.



		SELCO DIVISION - BP RESOURCES CANADA LIMITED
J & L PROJECT		
SHOWING # NS 3		
LOOKING TOWARDS 40°		
SCALE 1:25	DRAWN BY: N.H.	FIG. 13
DATE AUG. 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/BE	PROJ. 10115	
REPORT BPVR 84-53		



	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25118	<0.01	0.01	0.089	2.1	0.1	0.91
25119	0.06	<0.01	17.700	10.6	5.0	0.10
25120	<0.01	0.01	0.144	2.1	0.1	0.68

LOCATION: L 8+00N, O + 51.5 W

ELEVATION: 1270 m.

0 0.5 Metre

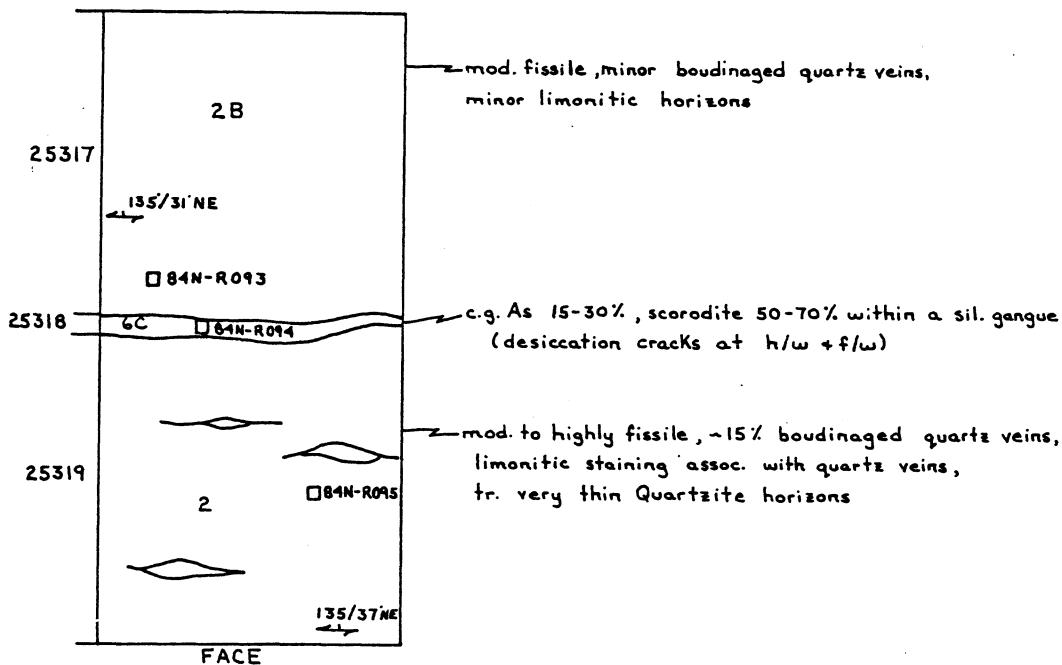


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J & L PROJECT

SHOWING # NS 9

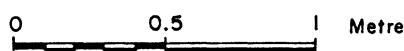
SCALE	1 : 25	DRAWN BY	N. H.	FIG.	14
DATE	AUG., 1984	DRAFTED BY	R. P.		
M.T.S.	82M/8E	PROJ	10115	REPORT	BPVR 84-53



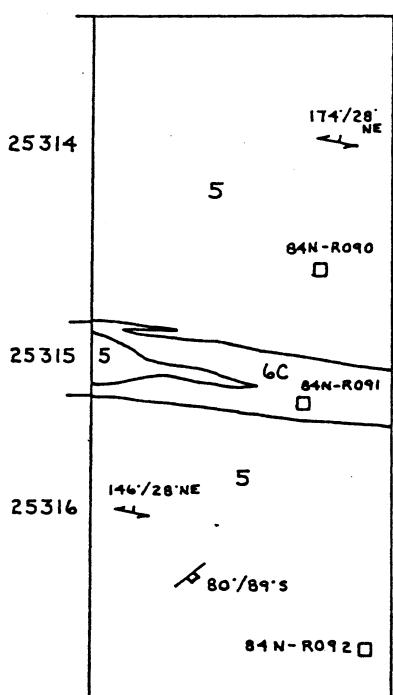
	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25317	<0.01	<0.01	0.072	0.3	0.2	0.71
25318	0.01	<0.01	18.200	0.3	3.3	0.05
25319	<0.01	<0.01	0.209	0.7	1.1	0.77

LOCATION: 45m. due west of LB+00 N,
0+27W

ELEVATION: 1270 m.



		SELCO DIVISION - BP RESOURCES CANADA LIMITED
J & L PROJECT SHOWING # NS II LOOKING TOWARDS 50°		
SCALE 1: 25	DRAWN BY: N.H.	FIG. 15
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/8E	PROJ. 10115	REPORT BPVR 84-53



clean, bleached, mod. limonite stain + minor jarosite, tr. Py
locally mod. fissile with ser. along foln.

bleached, clean, sheared + heavy jarosite stain

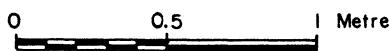
40% - 80% scorodite, 15-50% As + quartz gangue
+ 15% interfingering bleached, sheared Quartzite;
To west of sampled zone, M.S. grades into As +
scorodite stringers in sheared, bleached Quartzite

bleached (especially, near contact with M.S.),
minor intercalated ser. phyll., minor boudinaged quartz veins;
locally up to 5% limonite after Py

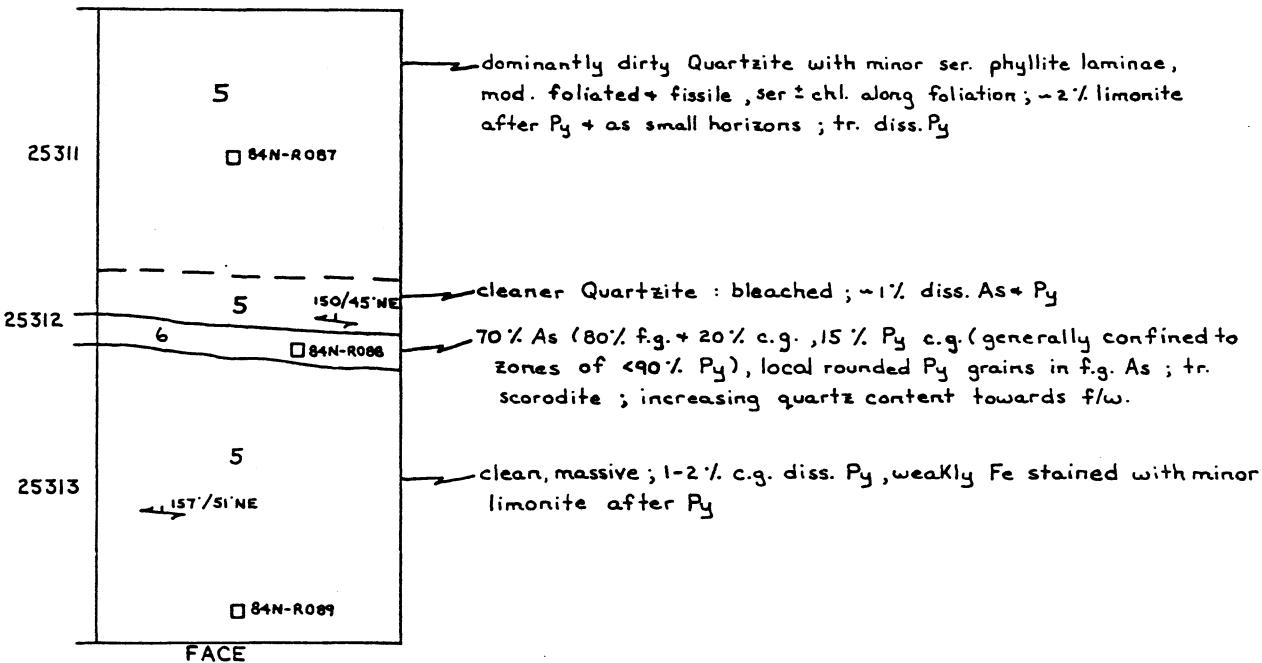
	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25314	<0.01	<0.01	0.104	0.3	<0.1	0.82
25315	0.08	<0.01	20.800	15.5	5.1	0.22
25316	<0.01	<0.01	0.236	0.3	0.2	0.73

NOTE: outcrop is slightly slumped

ELEVATION: 1280m.



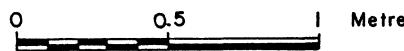
 SELCO DIVISION - BP RESOURCES CANADA LIMITED					
J & L PROJECT SHOWING # NS 13 LOOKING TOWARDS 54°					
SCALE	1 : 25	DRAWN BY	N. H.	FIG.	16
DATE	AUG., 1984	DRAFTED BY	R. P.		
M.T.S.	82 M/BE	PROJ.	10115	REPORT BPVR 84-53	



	%	%	%	g/t	g/t	m
	Pb	Zn	As	Ag	Au	L
25311	<0.01	<0.01	0.149	1.1	0.3	0.96
25312	0.08	<0.01	22.200	17.2	4.0	0.08
25313	<0.01	<0.01	0.578	0.5	0.2	0.84

LOCATION : 55 m. due west of B/L, 8+30N

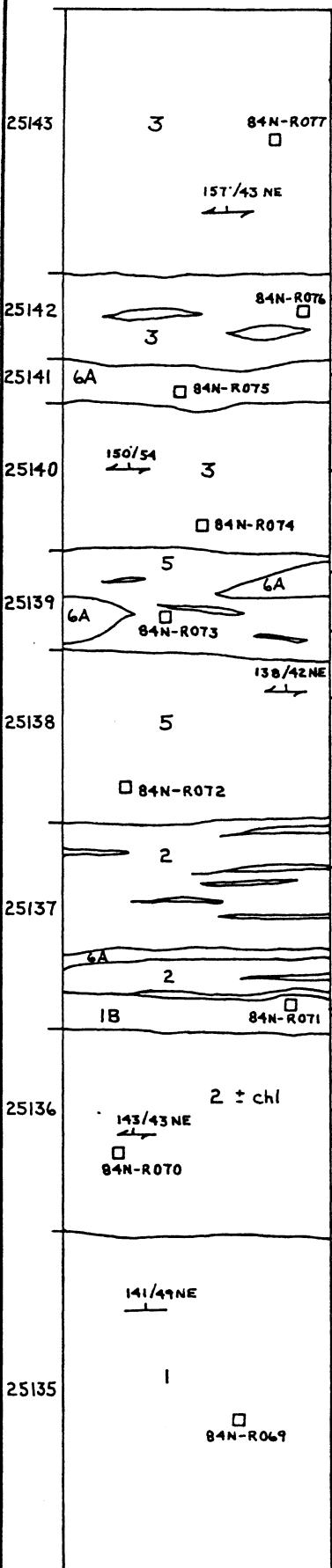
ELEVATION : 1280m.



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BP RESOURCES CANADA LIMITED

J & L PROJECT
SHOWING # NS 14
LOOKING TOWARDS 48°

SCALE 1:25	DRAWN BY: N.H.	FIG. 17
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82M/8E	PROJ. 10115	REPORT BPVR 84-53



25143 3 84N-R077 → narrow talc zones,
 heavy jarosite + limonite
 157/43 NE
 25142 3 84N-R076 → minor ZnS + PbS assoc. with
 stretched quartz lenses
 (ZONE 1)
 25141 6A 84N-R075 → strongly leached: <35% As, minor
 scorodite, Py, jarosite + chlorite
 (ZONE 2)
 25140 150/54 3 → minor limonite, jarosite +
 diss. Py
 84N-R074
 25139 5 6A 84N-R073 → larger pods contain > 90% As: smaller pods contain
 > 10% Py, minor jarosite + scorodite, tr. PbS + ZnS
 (ZONE 3)
 → highly bleached; dissem. sulphides
 138/42 NE

25138 5 → dirty; grades down into a Qtz-Ser Phyll
 minor f.g. Py stringers + jarosite
 tr. diss. f.g. Py
 84N-R072

25137 2 → M.S. bands (10-15%): As, Py ± ZnS;
 thickest band displays 'milled texture'
 + contains 75% As, 15% Py in a sil. matrix; (tr.)
 ZnS prominent in thinner, more Py-rich
 bands
 ZONE 4

25136 1B 84N-R071 → Dol. + Lst; tr. diss. f.g. As
 143/43 NE

25136 2 ± chl → minor boudinaged quartz veins
 + cb. f.f.
 tr. c.g. Py stringers

25135 1 84N-R069 → thin banded,
 small crenulations + minor
 folds + limonite staining

	% Pb	% Zn	% As	% Ag	% Au	m L
25135	< 0.01	0.01	0.022	2.1	0.1	0.93
25136	< 0.01	0.01	0.008	7.2	0.3	0.65
25137	0.2	0.03	5.670	23.3	4.1	0.66
25138	< 0.01	0.02	0.079	0.3	0.1	0.43
25139	0.32	0.02	3.800	32.2	3.4	0.31
25140	0.19	0.04	0.205	11.3	0.3	0.33
25141	0.59	0.03	15.500	56.8	15.9	0.16
25142	0.78	0.60	0.465	25.4	0.7	0.22
25143	0.02	0.02	0.059	0.3	0.1	0.86

LOCATION: 'Old Workings'; 115m. west of
L2+00N, 0+55 W

ELEVATION: 1030 m.

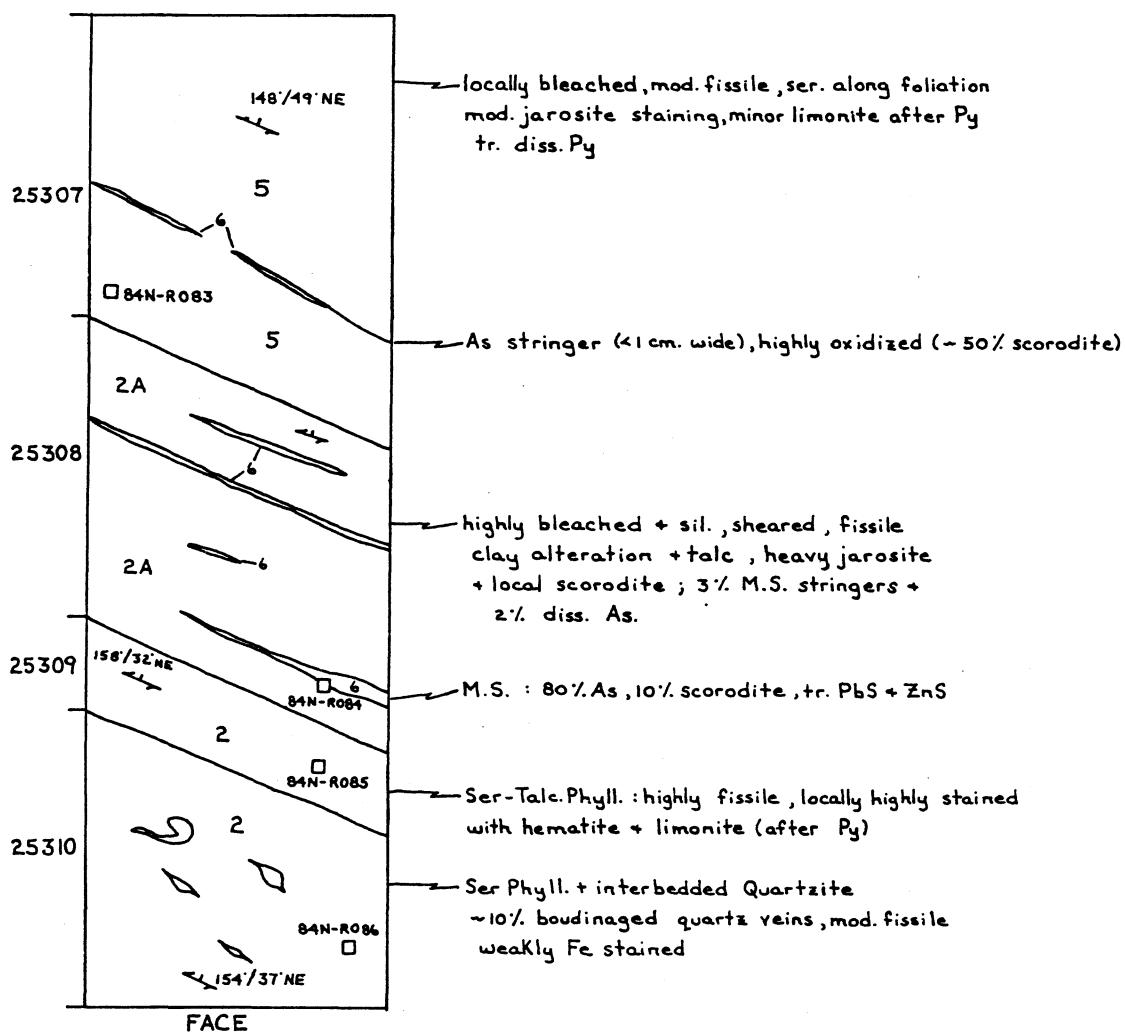


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J & L PROJECT SHOWING # NS 20

0 0.5 1 Metre

SCALE 1:25	DRAWN BY: N.H.	FIG. 18
DATE AUG., 1984	DRAFTED BY: R. P.	
N.T.S. 82 M/BE	PROJ. 10115	REPORT BPVR 84-53

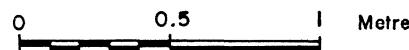


	%	%	%	g/t	g/t	m
	Pb	Zn	As	Ag	Au	L
25307	0.03	<0.01	0.075	1.1	0.3	0.70
25308	0.34	0.05	2.370	8.7	2.3	0.81
25309	0.43	0.03	1.280	16.8	1.4	0.28
25310	0.02	0.07	0.060	3.1	0.2	0.75

NOTE: showing offset 12m. at 124° from NS-20, face is on outcrop that has slumped slightly; very little offset because mineralization can be traced along strike past the slumped area.

LOCATION: 130 m. west of L2+00N, 0+55W

ELEVATION: 1040 m.

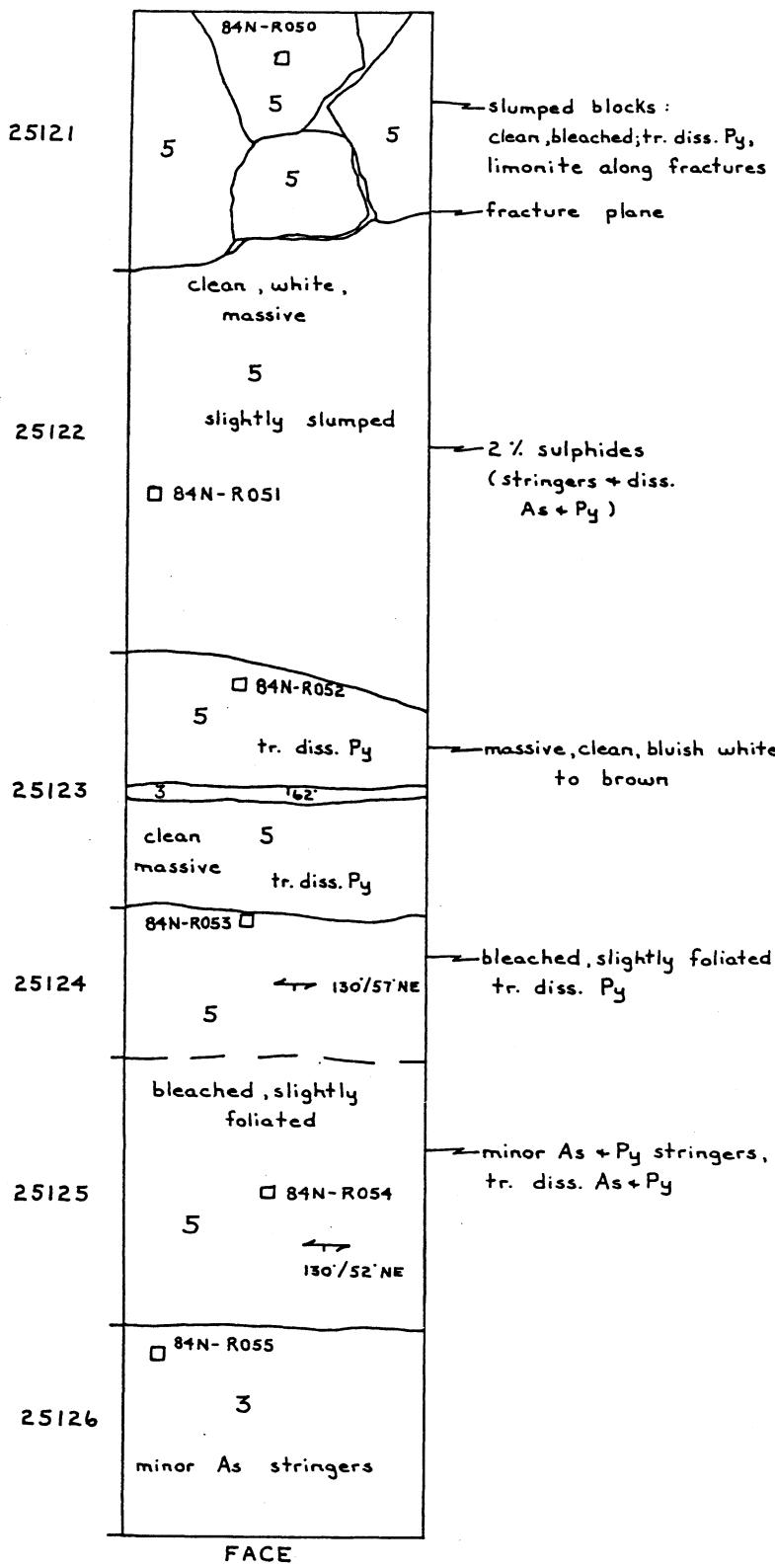


SELCO DIVISION -
BP RESOURCES CANADA LIMITED

J & L PROJECT

SHOWING # NS 21
LOOKING TOWARDS 48°

SCALE 1: 25	DRAWN BY: N.H.	FIG. 19
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/BE	PROJ. 10115	REPORT BPVR 84-53



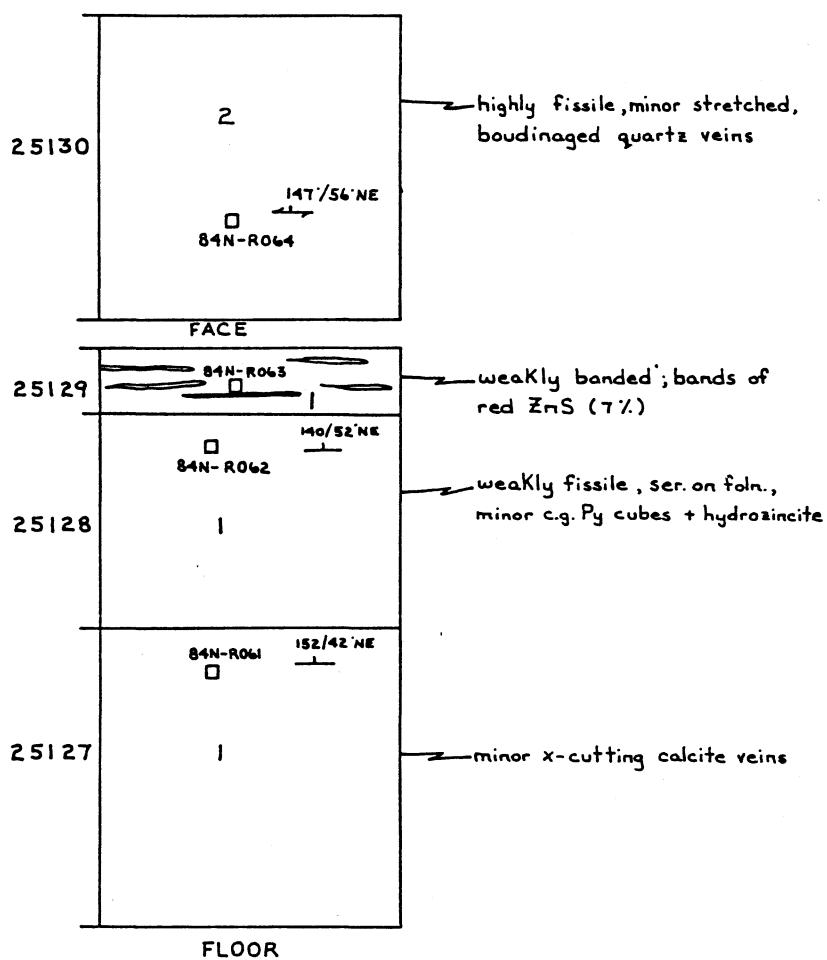
	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25121	<0.01	<0.01	0.425	4.1	0.1	0.54
25122	<0.01	<0.01	0.630	2.1	0.1	0.69
25123	<0.01	<0.01	0.287	0.3	<0.1	0.47
25124	<0.01	<0.01	0.032	1.4	<0.1	0.27
25125	<0.01	<0.01	0.010	0.3	<0.1	0.55
25126	<0.01	0.01	0.509	0.7	<0.1	0.43

LOCATION: L4+00N, 13.5 m. at 140° from 1+35W

ELEVATION: 1110 m.

0 0.5 | Metre

BP		SELCO DIVISION - BP RESOURCES CANADA LIMITED
J & L PROJECT SHOWING # NS 22		
SCALE 1: 25	DRAWN BY: N.H.	FIG. 20
DATE AUG. 1 1984	DRAFTED BY: R. P.	
M.T.S. 82 M/8E	PROJ. 10115	REPORT BPVR 84-53



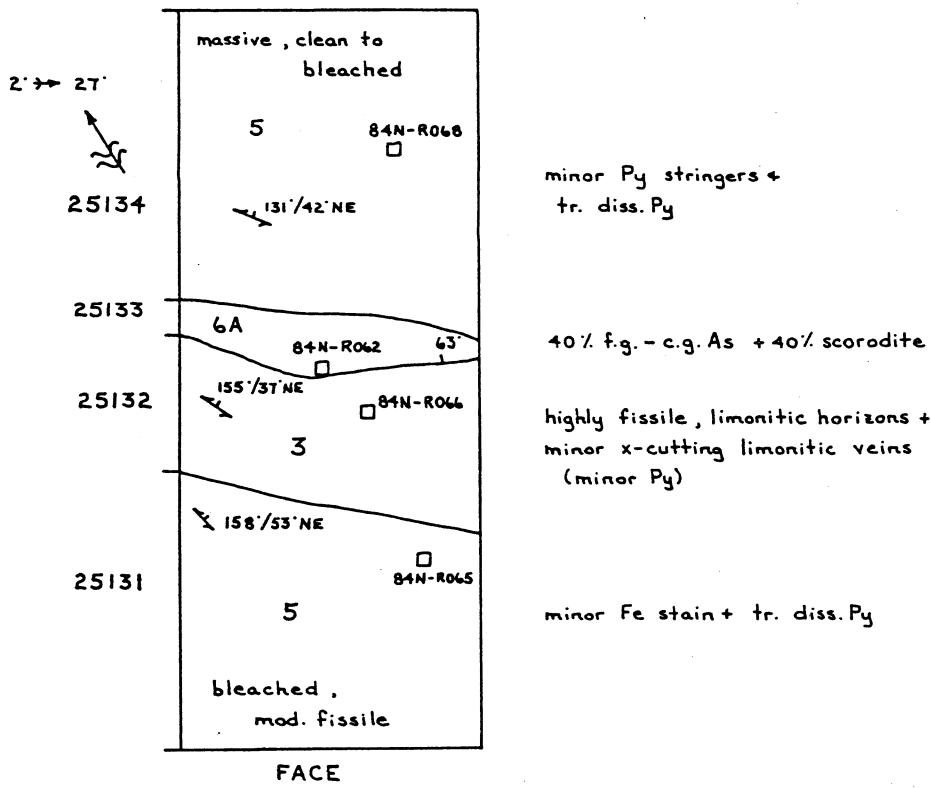
	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25127	0.04	0.02	0.009	1.0	<0.1	0.95
25128	0.09	0.13	0.096	1.4	<0.1	0.55
25129	0.16	3.58	0.148	0.7	0.2	0.20
25130	<0.01	0.04	0.006	0.7	0.3	0.95

LOCATION: 60 m. west of L4+00 N, 0+75 W

ELEVATION: 1070 m.

0 0.5 1 Metre

 SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT SHOWING # NS 24	
SCALE 1:25	DRAWN BY: N.H.
DATE AUG. 1 1984	DRAFTED BY: R.P.
N.T.S. 82 M/8E	PROJ. 10115
REPORT BPVR 84-53	



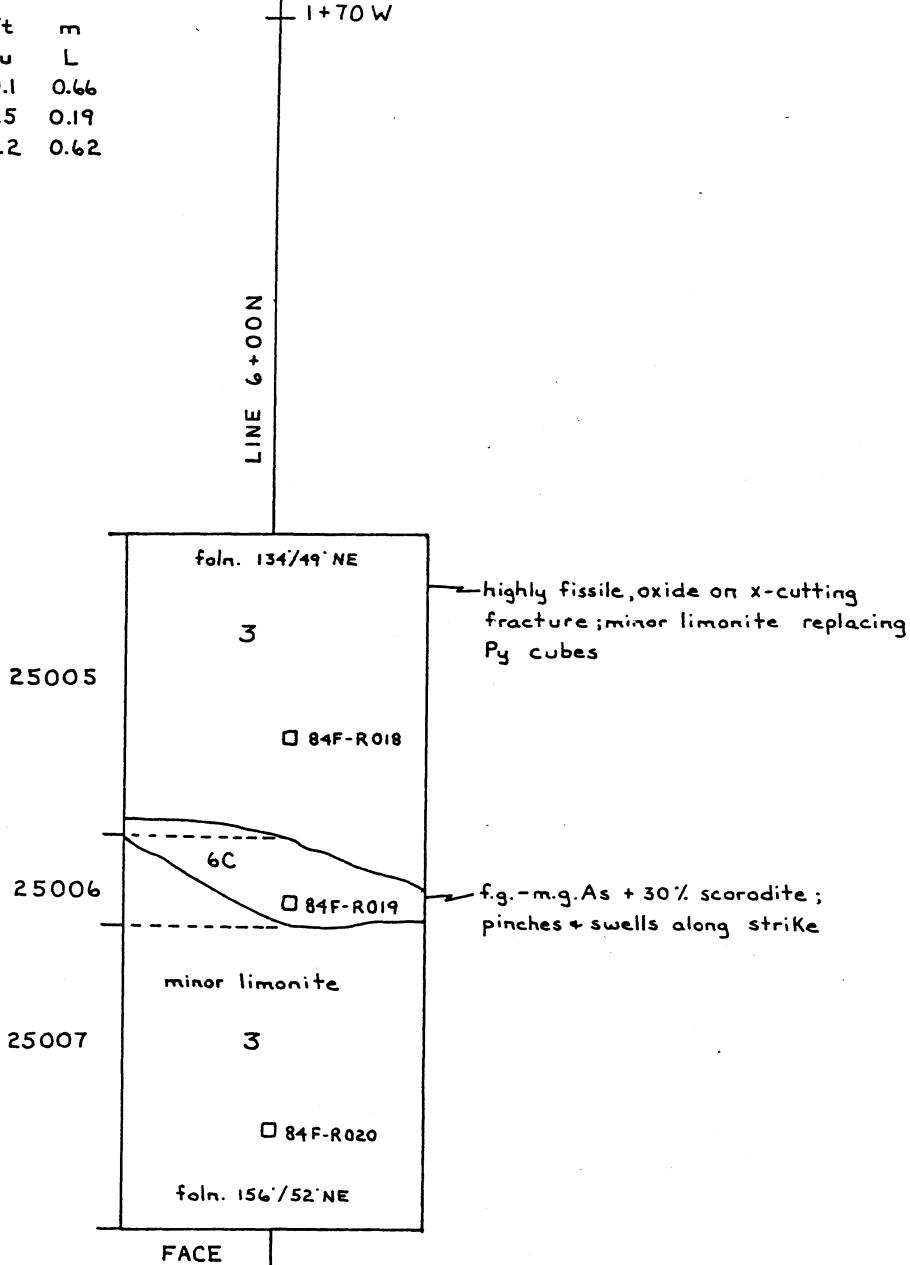
	Pb	Zn	As	Ag	Au	L
25131	<0.01	0.01	0.049	1.4	0.3	0.57
25132	<0.01	<0.01	0.055	0.7	1.7	0.27
25133	0.02	<0.01	19.900	1.4	0.6	0.14
25134	<0.01	<0.01	0.110	2.0	0.2	0.61

0 0.5 1 Metre

		SELCO DIVISION - BP RESOURCES CANADA LIMITED
J & L PROJECT SHOWING # NS 26		
SCALE 1 : 25	DRAWN BY: N. H.	FIG. 22
DATE AUG., 1984	DRAFTED BY: R. P.	
N.T.S. 82M/8E	PROJ. 10115	REPORT BPVR 84-53

	%	%	%	g/t	g/t	m
	Pb	Zn	As	Ag	Au	L
25005	0.10	0.35	0.052	4.8	0.1	0.66
25006	2.24	0.50	22.500	50.1	7.5	0.19

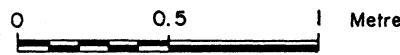
25007 0.06 0.02 0.299 1.7 0.2 0.62



LOCATION: L 6+00N, 1+71.6 W

ELEVATION: 1112 m.

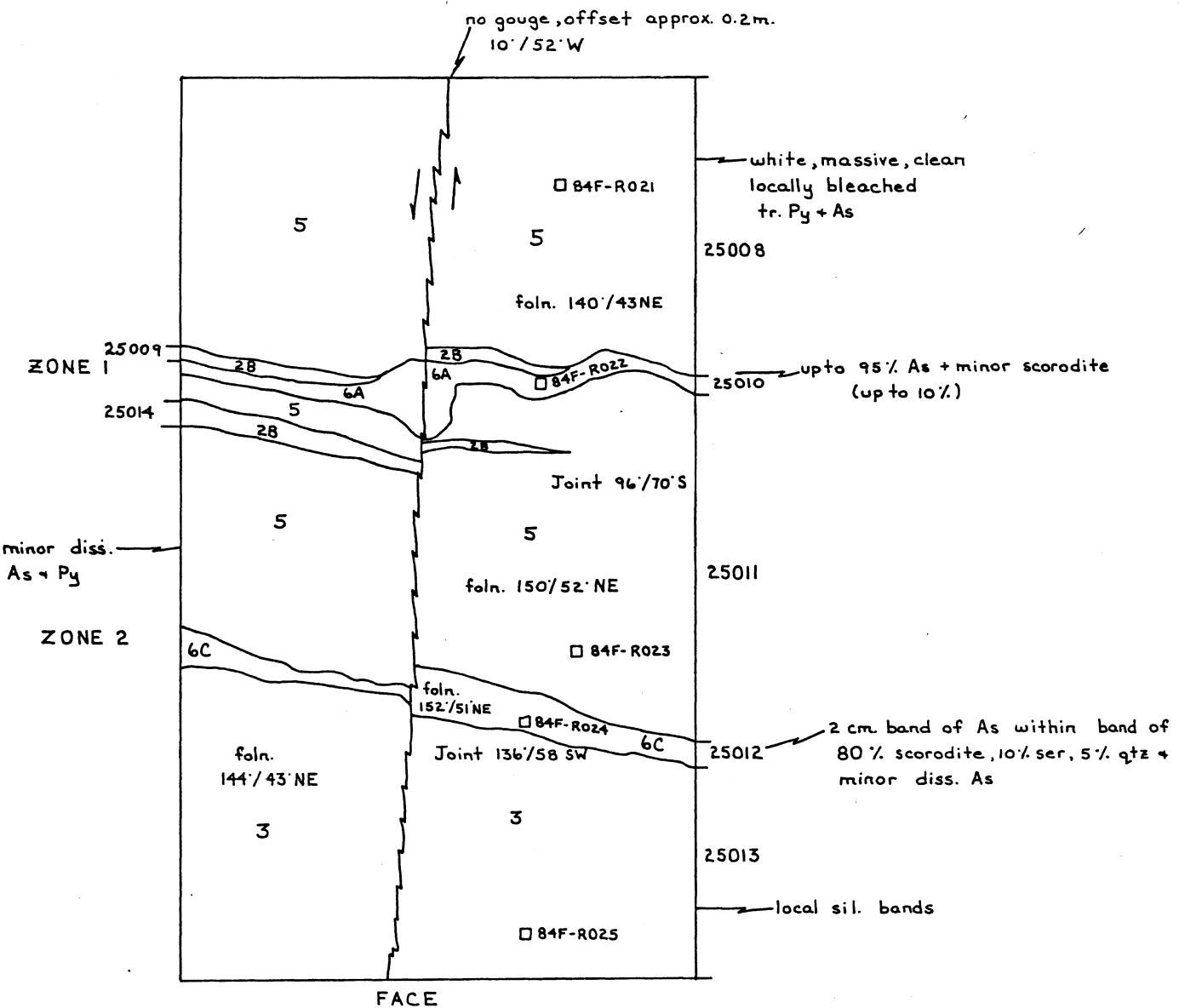
NOTE: another zone lies parallel to this showing, approx. 3.4 m. at a bearing of 210° and was sampled at a location of 12.57 m. at a bearing of 311° from L6+00N, 1+75W



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J & L PROJECT SHOWING# NS 28

SCALE 1: 25	DRAWN BY: F.T.M.	FIG. 23
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/8E	PROJ. 10115	
REPORT BPVR 84-53		

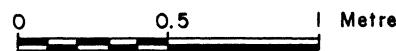


	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25008	<0.01	0.01	0.376	2.1	<0.1	0.66
25009	<0.01	0.01	0.216	1.0	0.3	0.11
25010	0.03	<0.01	25.100	3.8	5.3	0.17
25011	<0.01	<0.01	1.510	0.7	0.2	0.96
25012	<0.01	0.01	11.400	4.5	1.4	0.15
25013	<0.01	<0.01	0.171	0.7	<0.1	0.82
25014	<0.01	<0.01	0.684	0.3	<0.1	0.10

LOCATION: 12.57 m. from L6+00N, 1+75 W at a bearing of 311°

ELEVATION: 1125 m.

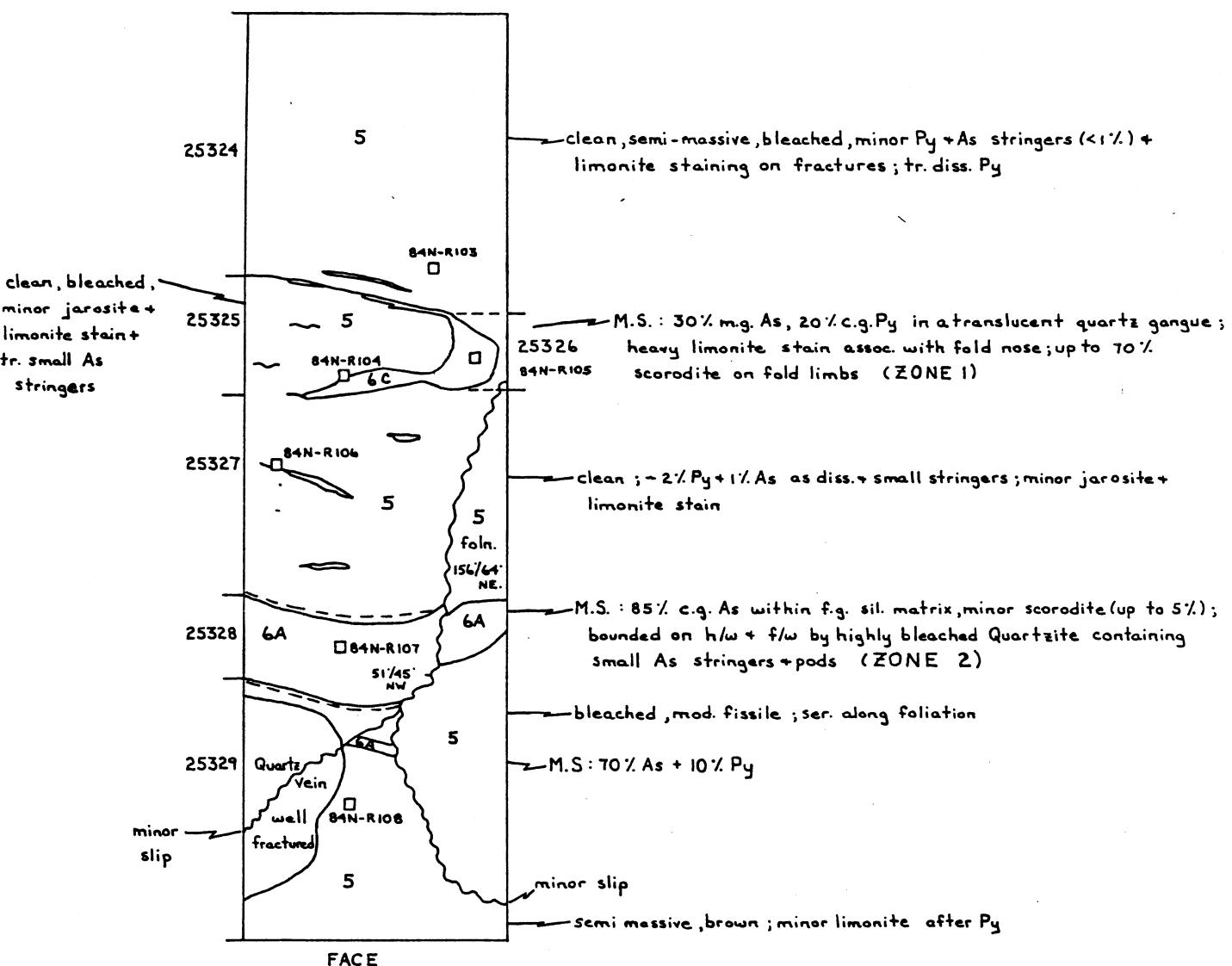
NOTE: another zone lies parallel to this showing, approx. 3.4 m. at a bearing of 37°. It was sampled on L6+00N at 1+71.6 W



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J & L PROJECT SHOWING # NS 30

SCALE 1:25	DRAWN BY: F.T.M.	FIG. 24
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/8E	PROJ. 10115	REPORT BPVR 84-53



	% Pb	% Zn	% As	g/t Ag	g/t Au	m L
25324	<0.01	<0.01	0.110	5.2	0.3	0.60
25325	<0.01	<0.01	5.870	10.8	5.0	0.41
25326	<0.01	<0.01	3.760	8.5	3.8	0.25
25327	<0.01	<0.01	0.207	1.5	0.5	0.53
25328	0.03	<0.01	15.500	0.3	4.6	0.18
25329	<0.01	0.01	1.040	0.3	0.3	0.77

LOCATION: 91m. at 142° from B/L, 12+00N.

ELEVATION : 1350 m.



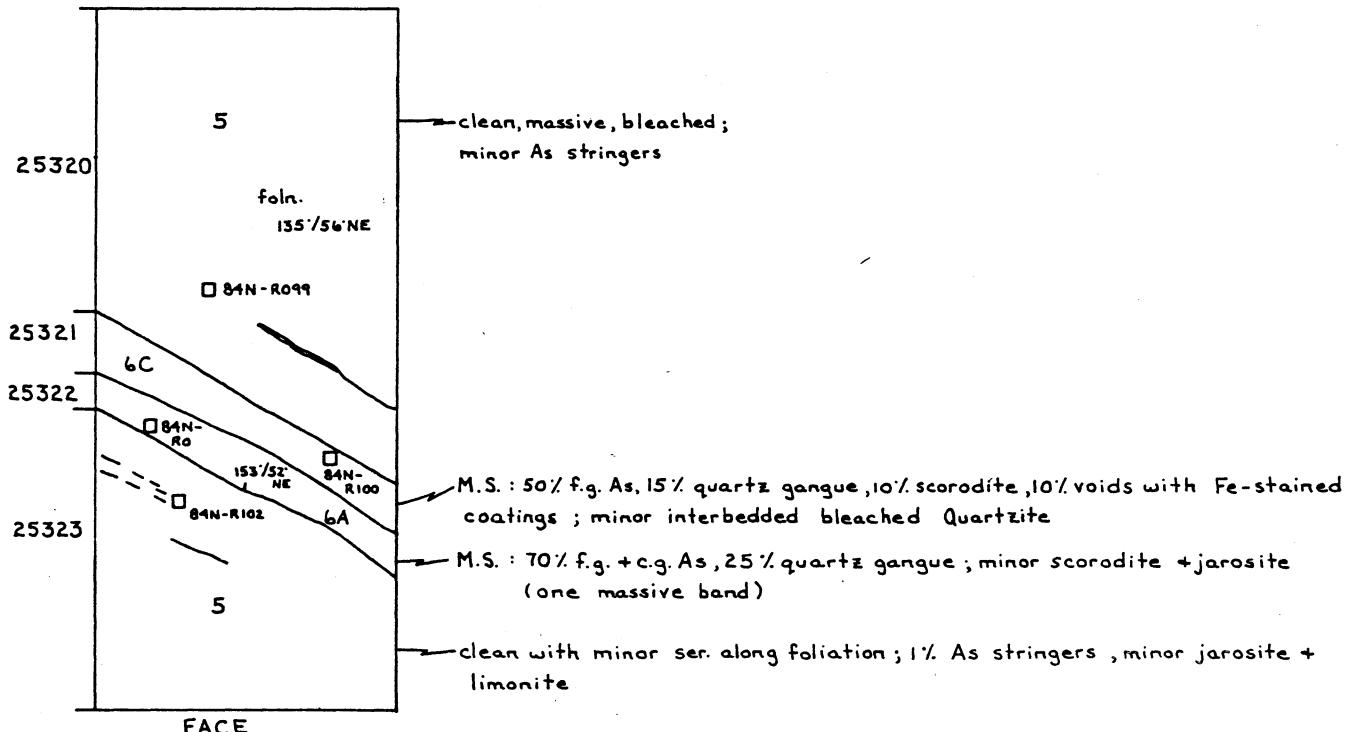
SELCO DIVISION -
BP RESOURCES CANADA LIMITED

J & L PROJECT

SHOWING # NS 37
LOOKING TOWARDS 43°

0 0.5 1 Metre

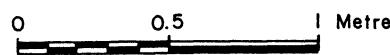
SCALE 1:25	DRAWN BY: N.H.	FIG. 25
DATE AUG., 1984	DRAFTED BY: R.P.	
N.T.S. 82 M/8E	PROJ. 10115	REPORT BPVR 84-53



	% Pb	% Zn	% As	% Ag	% Au	m.
25320	0.01	<0.01	0.266	2.4	0.3	0.88
25321	0.32	<0.01	23.300	18.0	4.6	0.10
25322	0.44	<0.01	26.700	55.0	4.0	0.09
25323	0.01	<0.01	0.629	4.5	0.3	0.74

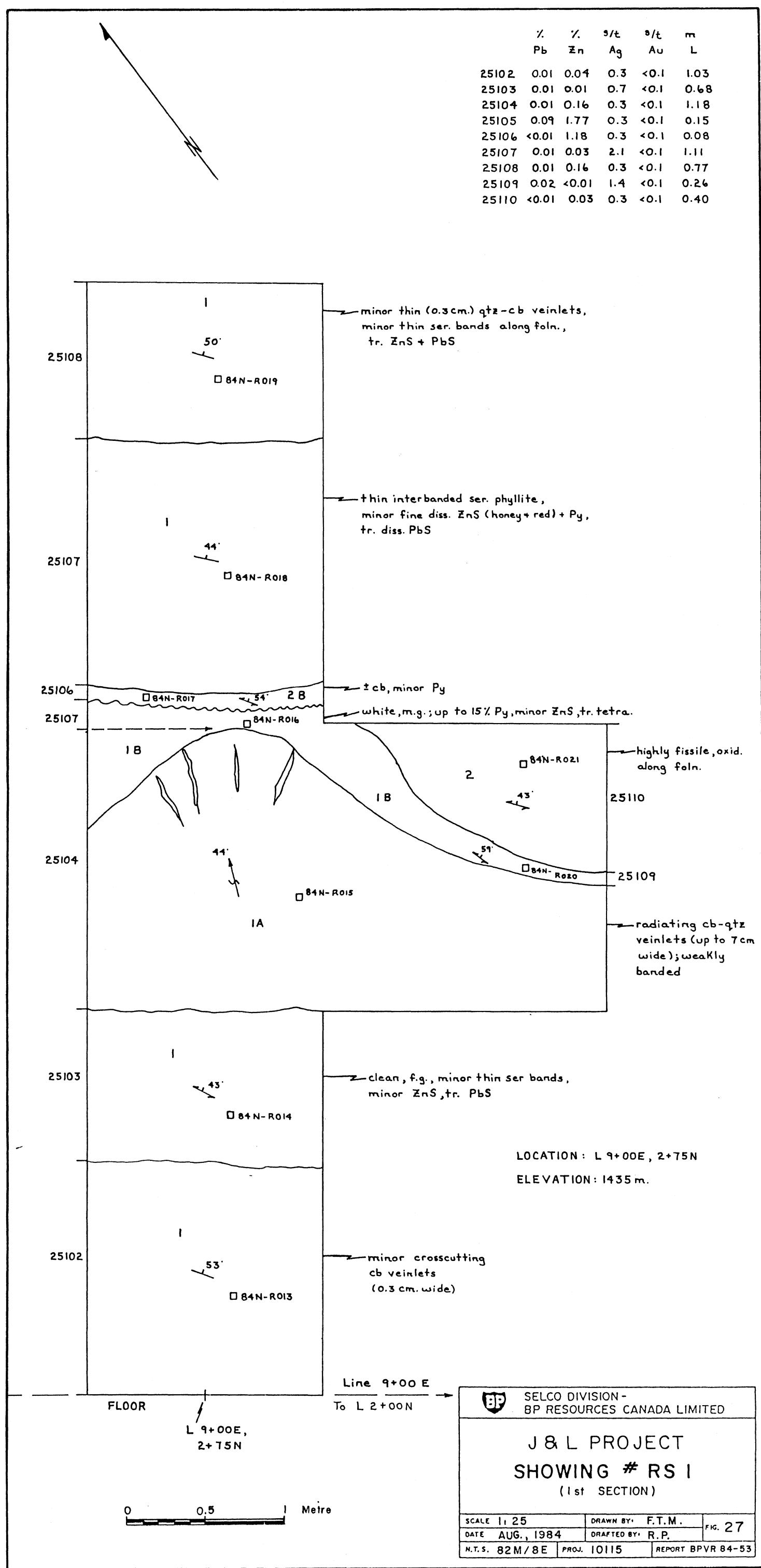
LOCATION: 53m. at 164° from B/L, 12+00 N

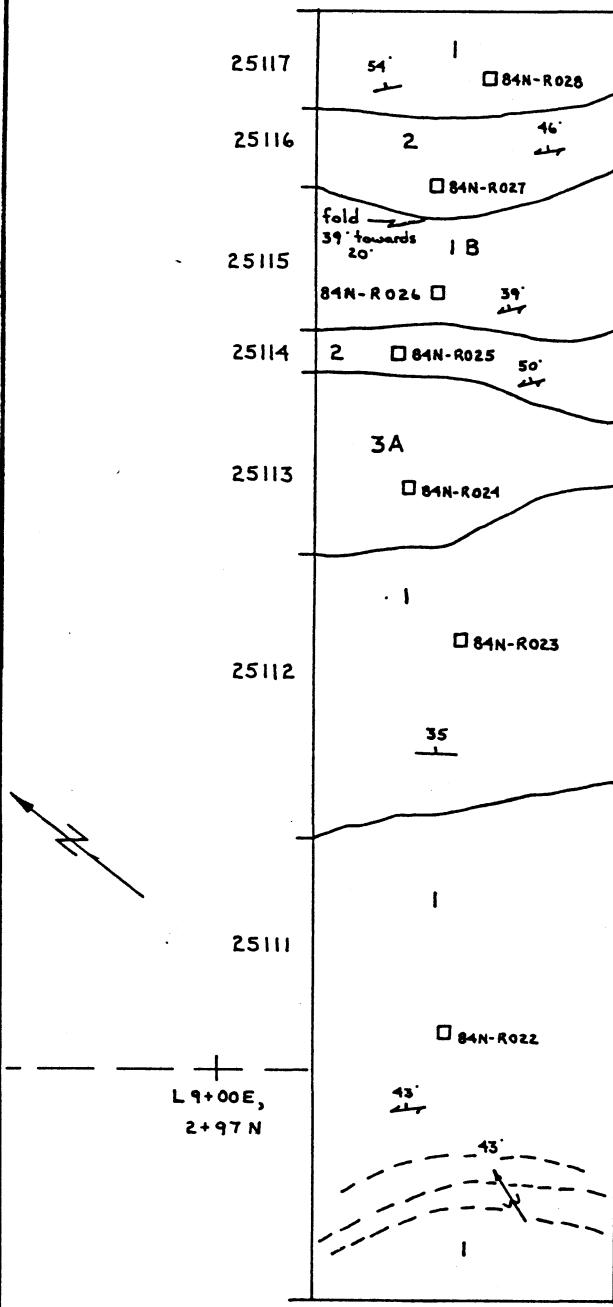
ELEVATION: 1370 m.



		SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT			
SHOWING # NS 38 LOOKING TOWARDS 16°			
SCALE 1: 25	DRAWN BY: N.H.	FIG. 26	
DATE AUG., 1984	DRAFTED BY: R.P.		
N.T.S. 82 M/8E	PROJ. 10115	REPORT BPVR 84-53	

	% Pb	% Zn	g/t Ag	g/t Au	m L
25102	0.01	0.04	0.3	<0.1	1.03
25103	0.01	0.01	0.7	<0.1	0.68
25104	0.01	0.16	0.3	<0.1	1.18
25105	0.09	1.77	0.3	<0.1	0.15
25106	<0.01	1.18	0.3	<0.1	0.08
25107	0.01	0.03	2.1	<0.1	1.11
25108	0.01	0.16	0.3	<0.1	0.77
25109	0.02	<0.01	1.4	<0.1	0.26
25110	<0.01	0.03	0.3	<0.1	0.40





	% Pb	% Zn	% Ag	% Au	m
25111	0.04	0.37	0.3	<0.1	1.09
25112	<0.01	0.02	0.3	<0.1	0.50
25113	<0.01	0.02	0.3	<0.1	0.55
25114	<0.01	0.03	1.4	<0.1	0.15
25115	0.02	1.22	0.3	<0.1	0.99
25116	<0.01	0.06	2.1	<0.1	0.19
25117	0.01	0.01	0.3	<0.1	0.49

LOCATION: L 9+00E, 2+96 N

ELEVATION: 1438 m.



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J & L PROJECT
SHOWING # RS 1
(2nd SECTION)

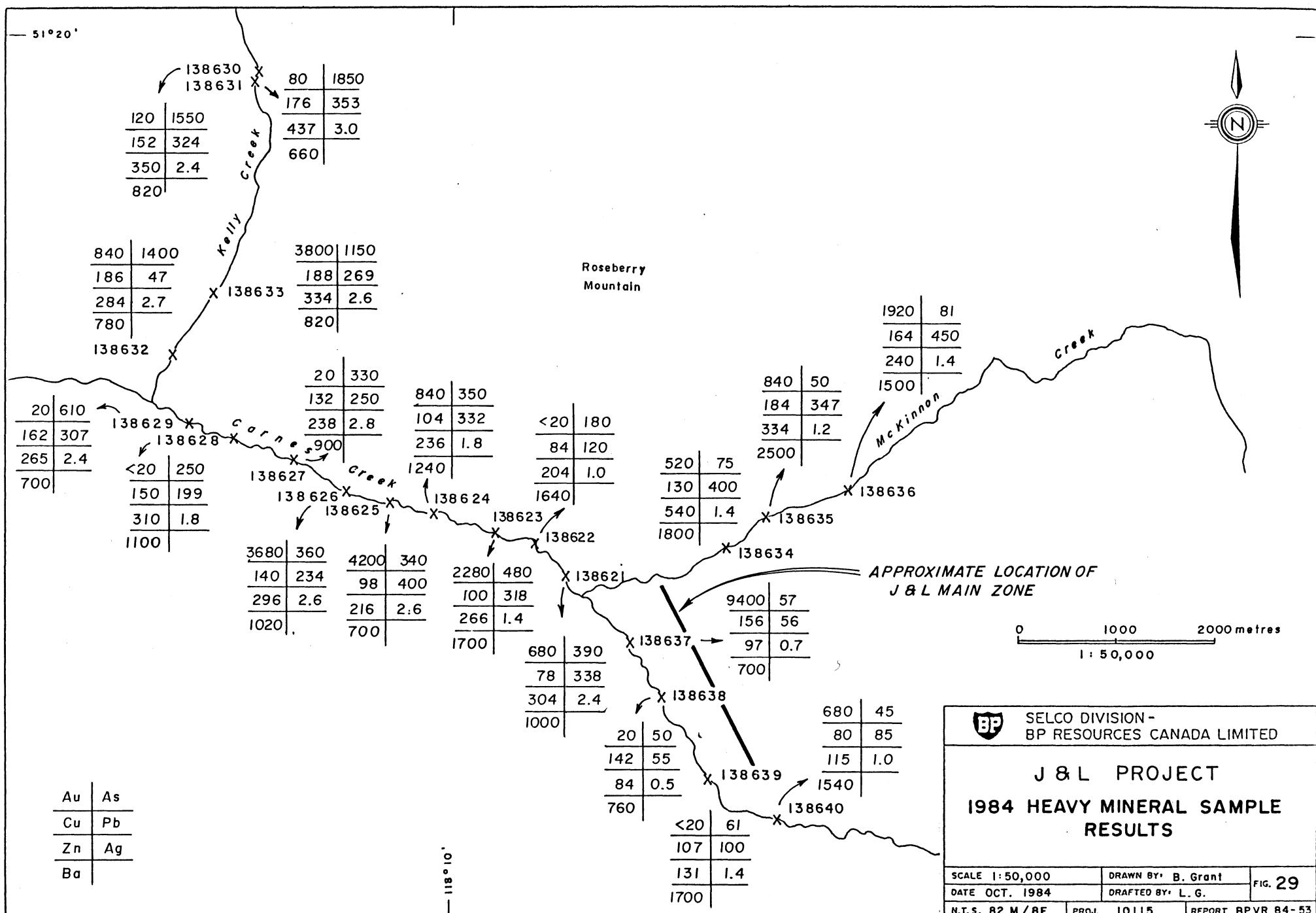
0 0.5 1 Metre

SCALE 1: 25	DRAWN BY: F. T. M.	FIG. 28
DATE AUG., 1984	DRAFTED BY: R. P.	
M.T.S. 82M/8E	PROJ. 10115	REPORT BPVR 84-53

occurrences within the Badshot limestones are shear related. These showings do not appear to be of any economic significance.

2. Heavy Mineral Sample Results

The results of the survey indicate that although the highest gold value occurs immediately below the showings (9400 ppb) the values then drop off significantly (20 to 700 ppb) for the first 1.5 km below the showings. The gold values then rise significantly (2000 to 4000 ppb) for the next two kilometres, or from 1.5 to 3.5 km downstream, after which gold values decrease to below anomalous levels. Arsenic shows a similar trend which is low below the main showing and increases to peak values about 1 km below the showings (480 ppm) then decreases slowly to the downstream end of the sampling. Although barium shows a similar trend with highs about 0.5 to 1 km downstream, a strong barium source is also noted upstream of the J & L (about 0.5 to 1 km). The base metals Pb-Zn-Ag show a peak similar to Au-As with high anomalous values at about 1 to 1.5 km below the actual showings. Copper shows an increasing trend with distance from the J & L Main Zone showings.



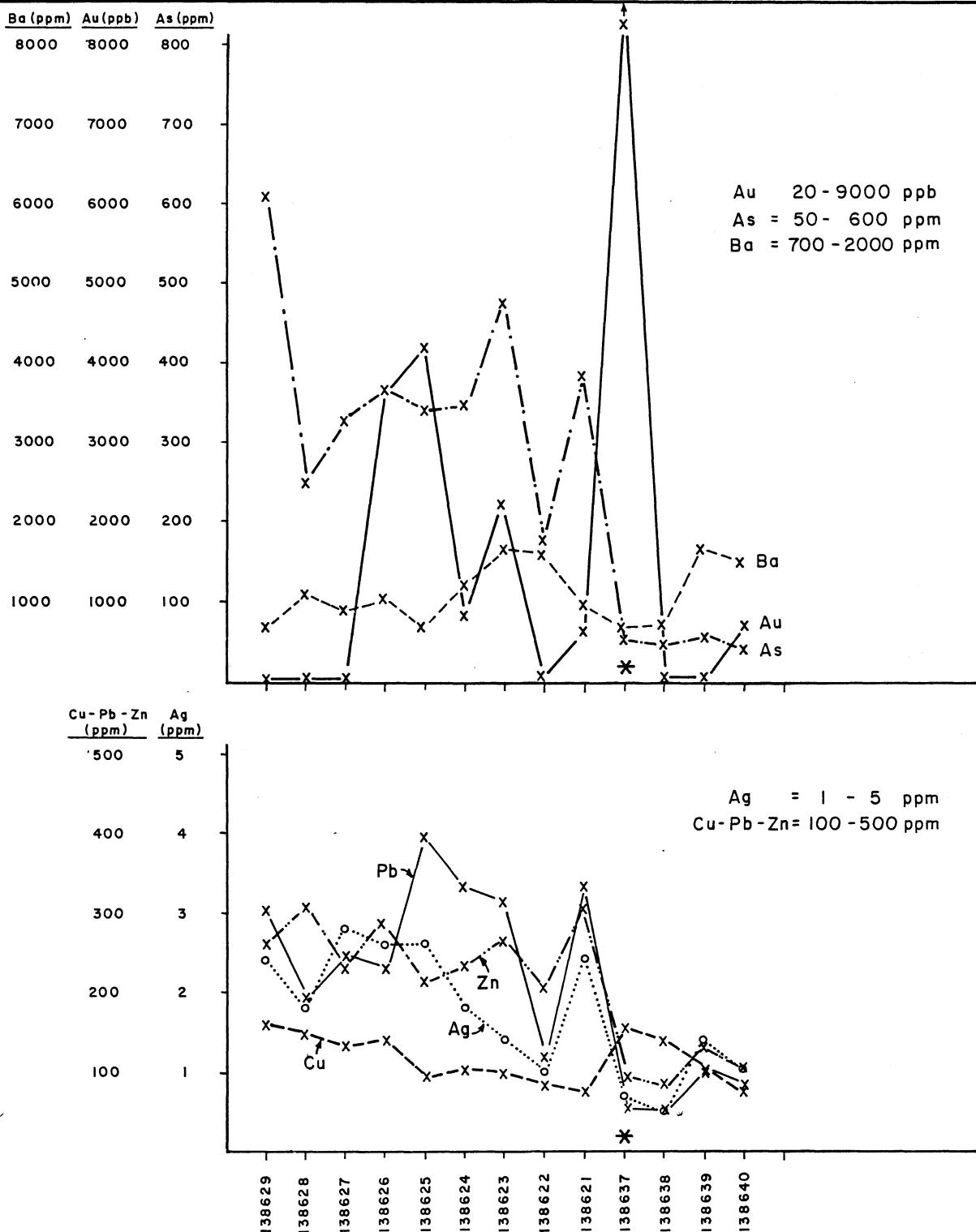
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BP RESOURCES CANADA LIMITED

J & L PROJECT
1984 HEAVY MINERAL SAMPLE
RESULTS

SCALE 1:50,000	DRAWN BY: B. Grant	FIG. 29
DATE OCT. 1984	DRAFTED BY: L. G.	
N.T.S. 82 M / BE	PROJ. 10115	REPORT BPVR 84-53

In interpreting these results, one must note that particularly with heavy minerals, sporadic highs and/or lows are to be expected in the results due to the errors introduced by sampling and concentrating. What is most importance are the general trends (See Figure 29A). All metals show a general trend to be relatively low (although still anomalous) immediately below the actual showings and the values increase downstream away from the showings. The values peak about 1 to 2 km from the source and then tend to decrease downstream. The reason for these trends are believed to be that material weathered from the showings is in the form of boulders and rock fragments. As they travel downstream they are eroded (milled) so that individual mineral grains are released at the 1 to 2 km distance which gives rise to the better heavy mineral results. As the material is moved further downstream, the grains are more finely milled and destroyed and mixed with greater amounts of unmineralized materials.

The Pb-Zn-Ag trends are closely parallel, indicating that these minerals are very closely associated and finely intermixed in the source material. Copper shows an increasing trend downstream and is not similar in pattern



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J & L PROJECT
HEAVY MINERAL
DISPERSION

SCALE As Shown	DRAWN BY: B. Grant	FIG. 29A
DATE Feb. 1985	DRAFTED BY: L.G.	
N.T.S.	PROJ. 10120	REPORT BPVR 84 - 53

to the Pb-Zn-Ag indicating that perhaps there is a weak source of copper downstream of the J & L. The As-Au trends are similar, but the arsenic peaks before the gold values, indicating that placer gold buildup first requires the breakdown of the arsenopyrite minerals.

The metal values (both base and precious metals) are anomalous for the full extent of the sample pattern, at a distance of 5 km from the J & L showings. Further sampling downstream would be necessary to define the full extent of the anomalous results although this would be difficult to carry out due to very rugged terrain and because the lower reaches of Carnes Creek are now flooded by the Revelstoke Dam.

The gold, arsenic, silver and zinc anomalies obtained from samples from Kelly Creek are possibly draining from the strike extension of the North Zone and/or the Roseberry Prospect, but the source cannot be determined at this time.

3. Underground Development Results

The chip sampling results from Main Zone indicates that it attains substantial thicknesses and grades along a large

TABLE IV**
MAIN ZONE (U/G): WEIGHTED AVERAGE GRADES

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,539.168E	6.80 1.21	13.38 2.37	4.491 0.856	196.5 35.4	6.8 1.4	0.55 2.38
10,541.725E	9.69 1.84	19.41 3.60	3.323 0.844	251.4 49.3	10.0 2.0	0.30 2.17
10,544.368E	2.81 1.74	6.04 3.59	2.536 1.850	106.5 66.4	5.5 3.6	1.28 2.15
10,547.354E	2.32 2.11	11.34 10.32	3.810 3.516	56.2 51.5	5.8 5.4	2.10 2.50
10,550.600E*	3.73 2.14 3.07	8.84 5.20 5.51	2.980 1.523 2.395	101.3 57.1 77.2	6.0 3.7 6.4	1.25 2.22 2.40
10,554.370E	0.87 0.74 0.60	3.84 2.29 2.34	0.987 1.718 1.120	23.5 19.9 15.9	2.4 2.4 2.0	1.05 1.60 2.45
10,558.386E	1.23 0.86	2.23 1.62	3.542 2.666	42.3 29.7	5.1 3.7	1.45 2.35
10,562.509E	1.61 0.96 0.64	0.73 0.42 0.66	9.833 7.162 3.724	87.6 51.2 32.0	7.0 4.4 2.5	0.40 0.75 1.17
10,567.095E	7.85 3.40	9.09 3.95	3.760 2.178	258.1 113.0	10.4 4.8	1.69 3.88
10,570.305E	8.12 1.50	17.60 3.07	5.270 5.801	341.1 66.6	15.3 3.6	0.30 2.38
10,574.505E*	7.25 3.19 0.63	17.39 7.53 1.06	8.891 8.540 6.860	188.2 82.5 26.7	11.8 5.5 1.5	0.30 0.65 2.75
10,578.375E	7.03 3.18 0.78	14.13 6.64 1.55	6.970 3.792 2.043	255.9 112.9 29.9	15.5 7.2 1.8	0.20 0.47 2.60

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,582.086E	7.26	9.29	11.000	144.3	19.3	0.15
	1.74	2.13	4.684	32.6	5.0	0.87
	1.49	1.66	5.570	29.0	4.3	1.40
10,586.727E	7.77	16.48	2.203	146.4	15.0	0.20
	1.74	9.87	2.091	39.2	3.9	0.90
	1.03	4.28	2.476	18.1	2.0	2.50
10,591.015E	1.69	2.26	8.122	38.9	11.2	1.70
	1.53	2.31	7.295	36.0	10.1	2.05
	1.51	2.00	7.250	34.7	10.1	2.10
	1.38	2.07	6.592	32.4	9.1	2.50
10,594.557E	1.21	1.88	6.500	22.9	4.2	2.22
10,597.697E	1.45	2.52	6.789	43.6	8.4	2.34
10,601.370E	1.10	1.91	7.754	40.6	13.4	2.20
	1.01	1.77	7.542	37.2	12.3	2.58
10,604.755E	1.01	2.15	7.819	40.1	11.0	2.90
10,607.930E*	0.50	0.58	8.926	6.0	6.4	1.55
	0.97	1.63	5.866	20.6	8.8	4.65
10,610.845E	0.96	0.65	8.155	22.7	4.9	2.65
10,613.898E	5.84	3.01	8.181	184.2	8.6	1.75
	5.33	2.77	7.644	167.8	8.0	2.13
10,617.634E	1.91	4.61	9.612	61.0	16.1	2.05
	1.66	3.80	7.714	52.1	12.9	2.55
10,621.246E	1.46	2.69	9.190	49.5	12.7	1.40
	1.12	1.99	6.390	36.3	8.9	2.20
10,624.461E	2.36	2.75	4.243	41.9	7.4	2.03
	2.28	2.65	4.082	40.1	7.1	2.15
10,628.022E	2.18	2.64	5.003	34.2	9.2	2.30
	1.34	2.45	4.638	31.6	8.5	2.70

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,631.385E	2.01 1.97	3.28 3.21	8.480 8.314	49.4 48.2	14.3 14.0	2.00 2.13
10,634.413E	1.96 1.44	3.56 2.63	5.146 3.894	48.6 35.7	9.6 7.2	1.65 2.46
10,637.722E*	2.27 1.90 2.53	5.33 4.45 4.22	7.039 5.920 3.775	102.8 86.1 80.3	16.0 13.4 10.3	1.76 2.30 4.35
10,641.129E	3.32 2.28	4.77 3.28	7.405 5.163	119.5 82.8	14.3 9.8	1.20 2.20
10,644.632E	3.14 1.88	7.33 4.35	3.642 2.346	154.7 98.3	11.2 6.8	1.10 2.01
10,648.292E	2.12	3.98	4.727	51.8	10.4	2.50
10,651.745E	0.66 0.62	3.11 2.89	3.164 2.944	20.7 19.3	10.6 9.9	2.55 2.85
10,655.297E	1.43 1.36	2.64 2.51	5.524 5.258	45.2 44.2	14.9 14.2	1.95 2.25
10,658.010E	2.70 2.44	3.42 3.09	4.614 4.160	78.8 71.5	7.1 6.3	2.00 2.33
10,661.804E	2.08 1.88	5.64 5.12	8.360 7.634	71.2 65.7	10.6 9.8	2.55 2.76
10,665.072E	4.46 4.19	11.97 11.24	8.601 8.127	128.2 120.8	17.1 16.1	2.20 2.50
10,668.479E	1.93	7.50	8.921	44.4	18.1	2.50
10,671.642E*	1.50 1.38 1.15	1.96 1.78 1.25	8.359 7.562 6.075	34.8 32.8 33.3	13.5 12.1 7.5	2.00 2.50 5.60
10,674.753E	0.86 0.82	1.52 1.44	14.650 14.012	25.2 23.9	14.4 13.7	2.10 2.30

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,677.737E	0.15	0.32	2.056	12.7	2.5	1.40
	0.29	0.50	1.956	15.1	2.5	1.95
	0.26	0.51	1.842	13.7	2.4	2.52
10,681.429E	0.37	0.28	2.851	7.1	3.0	1.75
10,684.811E	1.49	2.25	4.344	49.0	10.4	2.32
10,688.109E	2.21	3.74	4.239	69.2	4.3	0.80
10,690.857E	8.39	12.47	9.049	216.6	14.6	0.15
	2.36	3.54	2.510	64.7	4.4	0.77
10,694.004E*	8.12	12.64	12.004	239.3	14.2	0.15
	2.08	5.68	4.398	55.7	6.9	4.80
10,696.743E	8.03	12.00	8.438	218.3	9.3	0.20
	1.23	1.92	1.549	37.2	1.7	2.02
10,699.851E	1.33	2.00	4.514	37.9	6.2	1.30
	0.70	1.04	2.416	19.7	3.3	2.65
10,700.831E	1.14	4.25	14.346	49.7	15.9	1.45
	0.94	3.50	11.755	40.4	12.9	1.85
10,702.752E	1.10	4.55	8.967	37.0	5.8	0.75
	0.31	1.19	3.079	12.2	1.8	2.46
10,704.822E	0.09	0.24	1.730	5.8	1.4	1.90
10,709.119E	0.91	1.12	3.549	22.4	3.4	0.55
10,712.431E	3.25	5.15	12.048	81.1	18.0	0.25
10,714.999E	1.33	3.36	9.090	43.8	9.0	0.16
10,717.594E*	0.32	0.52	5.905	15.3	4.9	1.33
	0.52	0.74	3.487	19.3	2.8	3.80
10,721.641E	0.77	0.82	12.900	45.0	8.5	0.37
10,724.789E	0.27	0.07	15.351	11.4	4.5	0.85

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,727.988E	0.23	0.01	17.656	14.2	12.8	0.30
10,730.940E	0.60	0.85	8.385	20.7	4.8	1.40
	0.43	0.59	6.392	15.6	3.4	2.32
10,733.691E	1.55	2.56	5.659	33.3	4.9	2.25
	1.50	2.49	5.428	32.6	4.7	2.42
10,737.382E	3.17	4.60	8.430	58.3	7.5	1.75
	2.61	3.78	7.313	48.6	6.3	2.50
10,739.643E	3.05	4.69	5.920	62.4	9.0	1.90
	2.13	3.26	4.805	43.8	6.4	2.80
10,743.535E	5.59	8.86	3.143	108.3	6.0	2.15
	5.20	8.23	2.954	100.8	5.6	2.43
10,746.869E	6.11	7.62	6.699	101.8	7.9	1.60
	4.32	5.39	5.070	73.9	5.6	2.48
10,749.640E*	4.31	8.39	6.799	73.1	7.1	1.30
	2.40	4.64	3.945	41.5	4.1	2.47
	2.29	4.99	5.129	47.9	4.7	2.50
10,752.460E	5.02	5.21	3.997	95.5	4.4	1.00
	2.63	2.77	2.464	56.5	2.4	2.31
10,755.766E	6.88	10.60	4.788	138.9	8.0	1.55
	4.67	7.23	3.265	95.4	5.4	2.50
10,758.589E	7.84	9.36	4.495	150.2	7.9	1.73
	6.84	8.16	4.028	130.6	7.0	2.18
10,761.867E	9.33	9.36	2.652	180.9	4.9	1.60
	7.90	7.94	2.272	153.9	4.2	2.18
10,765.054E	8.45	8.46	2.281	179.7	4.2	1.75
	7.21	7.23	1.978	153.8	3.6	2.35
10,768.483E	3.77	3.56	1.644	80.1	2.0	2.10

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,771.600E	5.85	9.69	1.690	101.3	3.7	1.55
	5.45	9.43	1.542	94.0	3.5	1.80
	4.59	7.93	1.459	79.9	3.0	2.53
10,774.785E*	5.68	8.11	1.097	102.2	2.7	1.30
	3.83	5.52	0.940	70.2	1.9	2.20
	1.27	2.69	0.379	24.9	0.6	5.15
10,778.267E	5.27	10.94	1.048	93.9	3.1	1.50
	3.96	8.12	1.060	71.1	2.5	2.10
10,781.672E	4.63	8.16	1.022	95.3	1.7	2.27
10,784.742E	5.75	20.06	3.388	59.3	9.8	0.25
	1.33	8.06	2.932	12.5	1.9	1.34
	0.88	5.61	1.958	8.9	1.6	2.71
10,787.715E	1.74	6.38	2.685	31.2	1.8	0.78
	1.08	3.65	3.114	19.8	2.0	1.35
	0.78	2.57	2.129	13.3	1.5	2.52
10,790.594E	6.44	15.74	1.825	111.9	6.1	0.50
	0.98	2.47	1.225	17.7	1.4	2.20
10,793.277E	1.63	3.66	3.445	31.7	3.0	0.70
	1.39	3.75	2.881	26.5	2.5	1.03
	0.64	1.71	1.345	12.9	1.5	2.14
10,796.431E	2.02	4.31	3.691	35.9	4.0	1.10
	1.97	4.26	3.523	35.1	3.8	1.18
	1.07	2.31	2.006	19.6	2.3	2.20
10,799.397E*	6.05	7.62	2.597	121.3	13.1	1.15
	3.18	3.98	1.742	64.7	7.0	2.75
	0.51	1.55	1.096	10.9	4.5	6.10
10,802.993E	4.63	11.61	1.545	93.8	4.0	1.10
	2.61	6.57	0.885	53.7	2.3	2.41
10,806.226E	3.86	9.27	3.346	75.6	4.9	1.40
	2.79	6.69	2.438	54.7	3.5	2.45
10,809.555E	2.11	6.71	1.485	43.0	2.2	1.70
	1.79	5.59	1.678	36.3	2.0	2.26

TABLE IV
(continued)

Section	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,812.566E	1.23	3.42	1.340	23.5	1.7	1.65
	1.02	2.80	1.186	19.7	1.4	2.20
10,815.490E	0.58	1.86	1.169	8.9	0.9	1.80
	0.58	1.87	1.110	9.0	1.0	2.04
10,818.779E	2.11	9.01	1.904	40.3	2.7	1.50
	1.39	6.08	1.298	28.5	1.9	2.46
10,822.267E*	2.42	9.02	0.617	39.5	0.8	2.20
	3.19	8.52	0.696	44.1	0.9	2.40
	1.00	2.78	0.278	15.6	0.5	8.45
10,825.704E	1.18	3.53	0.956	19.3	0.5	2.20
10,829.142E	1.21	5.91	0.536	19.7	0.5	2.45
10,830.696E	1.43	20.51	0.275	25.6	0.8	2.50

*JV Section

TABLE IV
(continued)

From	To	Zones - Weighted Averages					Est. True Width(m)
		% Pb	% Zn	% As	g/t Ag	g/t Au	
539.168E	830.696E	2.00	3.91	4.052	48.8	5.6	2.32
OR							
		1.86	3.71	3.839	45.6	5.4	2.46
539.168E	674.753E	1.68	3.20	5.192	50.1	7.7	2.58
677.737E	690.857E	1.02	1.56	3.068	32.6	5.2	1.63
694.004E	700.831E	1.42	3.57	4.628	41.5	6.1	2.83
702.752E	730.940E	0.46	0.80	5.284	17.0	3.4	1.30
733.691E	765.054E	4.20	5.52	4.116	83.4	5.1	2.42
768.483E	830.696E	2.17	5.82	1.323	40.2	2.0	2.34
OR							
		1.63	4.71	1.097	30.3	1.8	2.93
591.015E	674.753E	1.72	3.09	6.266	50.5	9.8	2.70
539.168E	700.831E	1.61	3.12	4.999	48.1	7.4	2.51
733.691E	830.696E	2.89	5.71	2.306	55.4	3.1	2.37
OR							
		2.41	4.95	2.010	46.4	2.8	2.75

**Please note that the weighted average grades were calculated over the same massive sulphide sections, but utilized various widths.

portion of the 1984 drifting. Since in many cases, the drift did not intersect the entire lateral extent of the mineralized zone, the short J.V. holes were drilled into both walls of the adit (See Plans 11-21). It should be remembered that the 1984 drifting started in the hangingwall stratigraphy, thus a gap in mineralization between 1983 and 1984 drifting.

Chip sampling and drilling in the 1984 drifting revealed 2 mineralized zones which erratically pinch and swell (See summary at end of Table IV). Zone 1 is 161.663 metres long (10539.168E-10700.831E), 2.51 metres wide and averages 1.61% Pb, 3.12% Zn, 4.999% As, 48.1 g/t Ag, 7.4 g/t Au. This is separated by a lower grade narrower section which is 28.188 metres long (10702.752E-10730.940E), 1.30 metres wide and averages 0.46% Pb, 0.80% Zn, 5.284% As, 17.0 g/t Ag and 3.4 g/t Au. Zone 2, which is 97.005 metres long (10733.691E-10830.696E), 2.37 metres wide and averages 2.89% Pb, 5.71% Zn, 2.306% As, 55.4 g/t Ag and 3.1 g/t Au, is "open" to the east.

As was mentioned previously, the drift did not always intersect the entire width of the mineralized sections,

therefore the above grades and thicknesses maybe, in part, misleading. Thus, a more realistic appraisal of the J & L Main Zone maybe indicated by the results from the 830 metre level drilling (J.V. holes) and their corresponding drift faces. This sampling indicates that the Main Zone between 10550.600E and 10822.267E (271.667 m) has a width of 4.05 m and an average of 1.48% Pb, 3.04% Zn, 3.592% As, 35.2 g/t Ag and 5.3 g/t Au (See Table VI). It also indicates that there are 2 separate mineralized zones. Zone 1 is 143.404 metres long (10550.600E-10694.004E), 4.09 metres wide and averages 1.67% Pb, 3.11% Zn, 5.028% As, 43.6 g/t Ag and 7.3 g/t Au. Zone 2 is 72.627 metres long (10749.640E-10822.267E) and open along strike, 4.04 metres wide and averages 1.43% Pb, 3.48% Zn, 1.432% As, 26.0 g/t Ag and 2.8 g/t Au.

It should be noted that the gold and arsenic grades tend to drop off from, approximately, 10768E to the end of the present drift where the zinc content has substantially increased. It should also be remembered that the grade calculations utilized uncut assays and that results were rounded off (See Plan 22).

TABLE V**
J.V. SECTIONS - WEIGHTED AVERAGE GRADES

From	To	Length (m)	% Pb	% Zn	% As	g/t Ag	g/t Au	Est. True Width (m)
10,550.600E	10,694.004E	143.404	1.67	3.11	5.028	43.6	7.3	4.09
10,749.640E	10,822.267E	72.627	2.24	4.51	1.659	41.5	2.8	3.20
			1.43	3.48	1.432	26.0	2.8	4.04
			1.56	3.22	1.163	29.6	2.0	4.71
			1.07	2.67	1.072	20.1	2.1	5.55
10,550.600E	10,822.267E	271.667	1.74	3.33	3.838	40.7	5.5	3.74
			1.48	3.04	3.592	35.2	5.3	4.05
			1.54	2.96	3.360	36.1	4.9	4.29
			1.32	2.74	3.175	31.5	4.7	4.60

**Please note that the weighted average grades were calculated over the same massive sulphide sections, but utilized various widths.

The thickest sections were found at 10799.397E (6.10 m) and 10822.267E (8.45 m) and the thinnest at 10714.99E (0.16 m), see Table IV. Again there appears to be a strong correlation between the arsenic and gold and between the lead and the silver. It also appears that the lead and zinc belong to a completely different population than the gold and arsenic. There is apparent, relative lack of gold associated with the coarse-grained arsenopyrite, but further work will be required to confirm this.

Chip sampling results from the two crosscuts indicate that the 10670E crosscut stratigraphy is relatively barren of mineralization and that there are 3 narrow mineralized sections within the 10820E crosscut, but they are not of economic importance.

Specific gravity readings reveal the following average specific gravities for the different stratigraphic units. Carbonaceous limestone (02) is 2.81, grey banded limestone (03) is 2.82, sericitic phyllites (04) are 2.91, chloritic phyllites (05) are 2.84, quartzites (06) are 2.92, quartz-sericite phyllite is 2.92, the quartz lens (13) is 2.85 and the massive sulphides (12) are 4.08.

Preliminary X.R.F. analyses indicate appreciable gallium

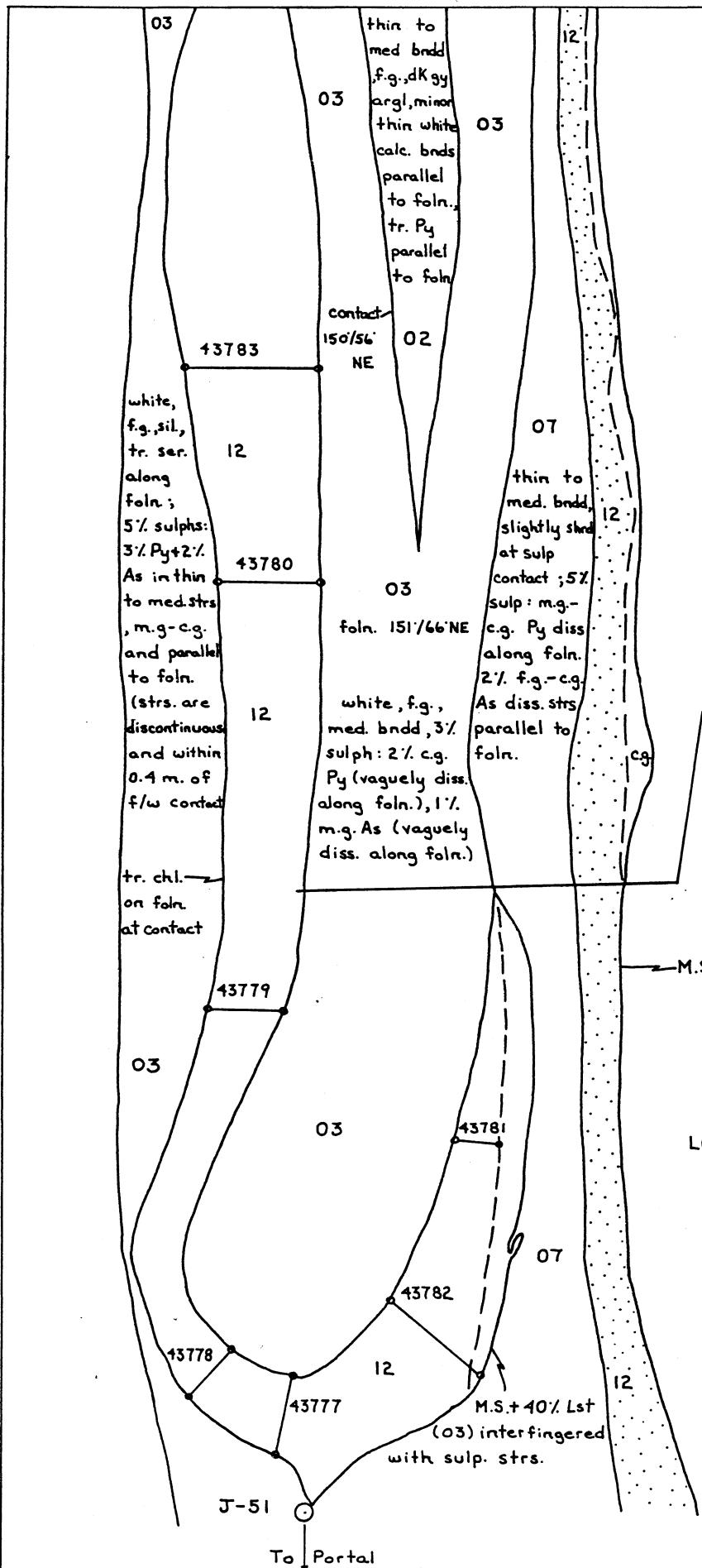
(up to 41 ppm) and germanium (up to 46 ppm) contents, but the significance of these elements is still to be determined. It also indicates erratic, high (up to 1.12%) copper values with a general, corresponding antimony high (up to 9825 ppm). Some of the copper can be explained by the visible secondary chalcopyrite, but the other copper highs and their corresponding antimony indicates the probable presence of tetrahedrite which has only been identified at a few locations. Thus, the Pb-Sb sulphosalts probably only account for a portion of the antimony content of the Main Zone. This should be studied further.

Preliminary isotopic analysis of coarse-grained galena from the 830 m drift give the following isotopic ratios:
206/204: 18.866, 18.875; 207/204: 15.705, 15.713; 208/204: 38.790, 38.826. McClay states that this indicates probable Mesozoic remobilization or introduction of lead into the ore system, but further work is required.

Chip sampling (See Figures 30 and 31) of folded massive sulphide bands in the 830 m drift was completed in order to investigate possible precious metal concentrations in various parts of a fold nose, but results were inconclusive.

J & L LEGEND

- 01 Graphite Schist
- 02 Carbonaceous Limestone and Gert
- 03 Limestone
- 04 Sericite (Muscovite), Sericite-Quartz Phyllite
(maybe calcareous)
- 05 Chlorite, Chlorite-Sericite, Chlorite-Quartz Schist
(maybe calcareous)
- 06 Quartzite, Quartz-Chlorite Schist (maybe calcareous)
- 07 Quartz-Sericite Schist (\pm minor chlorite)
- 08 Chert
- 09 Injected Quartz Breccia
- 10 Mylonite (sheared schist)
- 11 Calcite, Barite, Dolomite
- 12 Massive Sulphides
- 13 Quartz
- 14 Porphyry Dyke



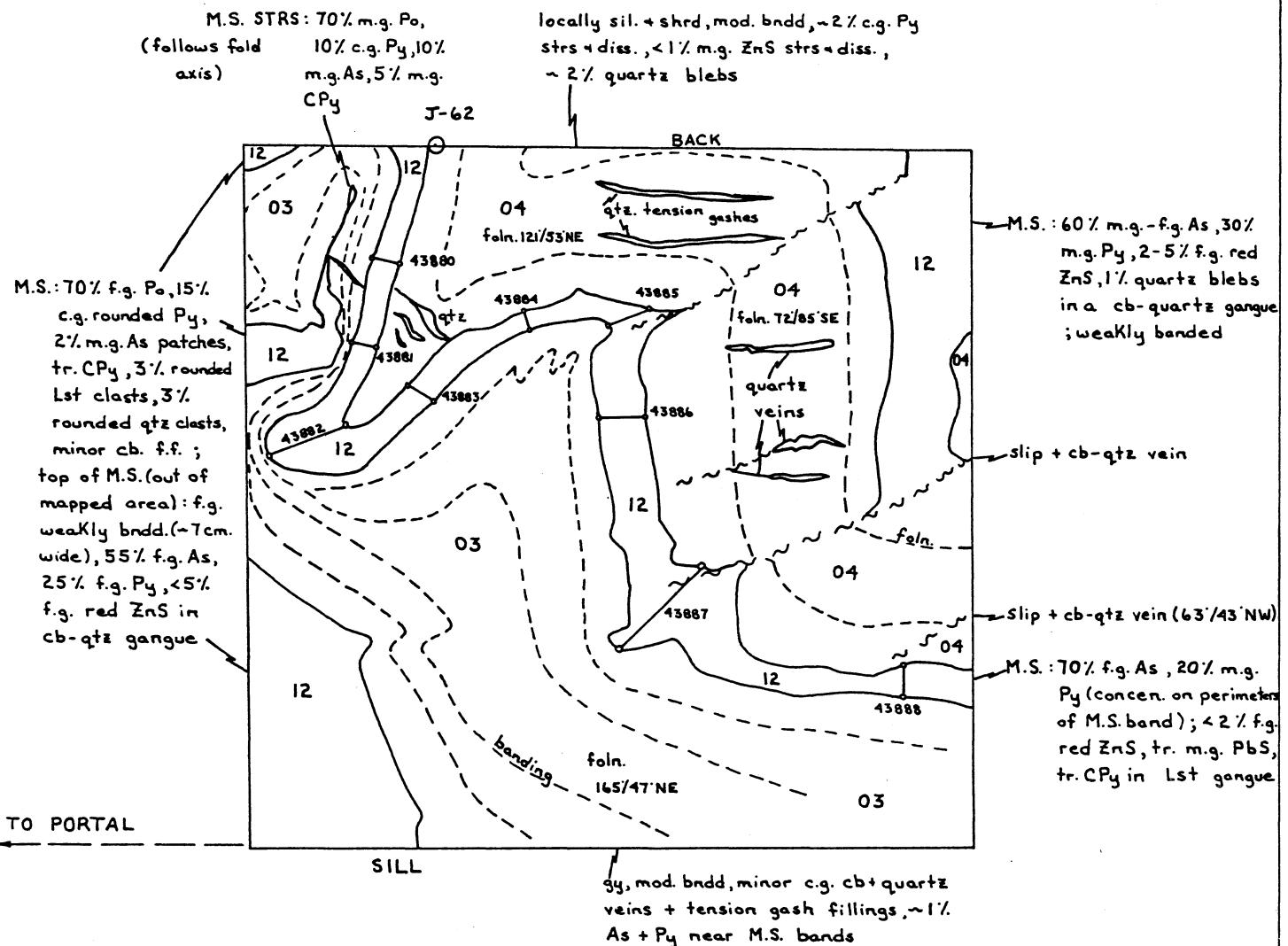
	% Pb	% Zn	% As	s/t Ag	s/t Au	m L
43777	0.12	0.01	25.800	30.9	2.7	0.30
43778	0.40	0.01	27.900	76.2	2.6	0.24
43779	0.06	0.01	13.700	23.5	1.2	0.30
43780	0.01	<0.01	5.590	4.1	1.4	0.40
43781	0.05	<0.01	28.900	7.9	3.1	0.20
43782	0.15	0.01	25.900	48.3	3.1	0.46
43783	0.01	0.01	6.730	1.6	1.1	0.52

M.S.: at tip of fold nose; 50% m.g.-c.g. As + 50% m.g.-c.g. Py (Py at outer half of band); South Limb: 80% m.g. As, 15% c.g. Py, 5% Lst North Limb: at J-51+2m: 50% As + 50% Py (inner half As-rich and interband with Lst, outer half c.g. Py); at J-51+3m: crosscutting Po + CPy veinlets at h/w contact; at J-51+4.5m: 45% Py, 25% Po, 20% As, 5% CPy, 5% Qtz-Ser, tr. Bornite+chl

M.S.: mainly f.g., milled texture; 50% As, 30% Py, up to 20% ZnS; m.g. to c.g. margin at f/w contact (As-rich + minor Py)

LOCATION: J-51 (10,572.062 E, 9950.637 N)

		SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT FOLD SAMPLING			
SCALE 1:25	DRAWN BY: F.T.M.	FIG. 30	
DATE AUGUST 1984		DRAFTED BY: R.P.	
N.T.S. 62 M/8E	PROJ. 10115	REPORT BPVR 84-53	

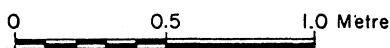


	% Pb	% Zn	% As	s/t Ag	s/t Au	m L
43880	1.30	1.83	11.900	33.2	14.8	0.10
43881	1.39	1.54	17.900	36.3	20.6	0.08
43882	0.98	0.57	12.500	26.1	14.3	0.32
43883	1.06	0.22	23.000	30.8	35.0	0.10
43884	3.20	0.90	18.300	66.5	20.6	0.06
43885	0.90	0.61	7.910	19.8	4.9	0.15
43886	1.38	3.47	8.950	50.6	9.7	0.85
43887	1.83	2.06	10.600	40.2	11.2	0.42
43888	0.79	3.84	10.300	26.2	13.6	0.12

LOCATION: J-62 (10,675.718E, 9925.993N.)
north wall

BP SELCO DIVISION -
BP RESOURCES CANADA LIMITED

J & L PROJECT FOLD SAMPLING



SCALE 1:25	DRAWN BY: F.T.M.	FIG. 31
DATE AUGUST 1984	DRAFTED BY: R.P.	
N.T.S. 82 M / 8 E	PROJ. 10115	REPORT BPVR 84-53

Duplicate sample results are generally within 20%, but there are some large discrepancies, but these can be expected with the rush duplicate chip sampling. Differences in the gold values can be explained with discrepancies in the arsenic content as can the silver values with the lead content. Check sample results appear fairly consistent with differences less than 18%.

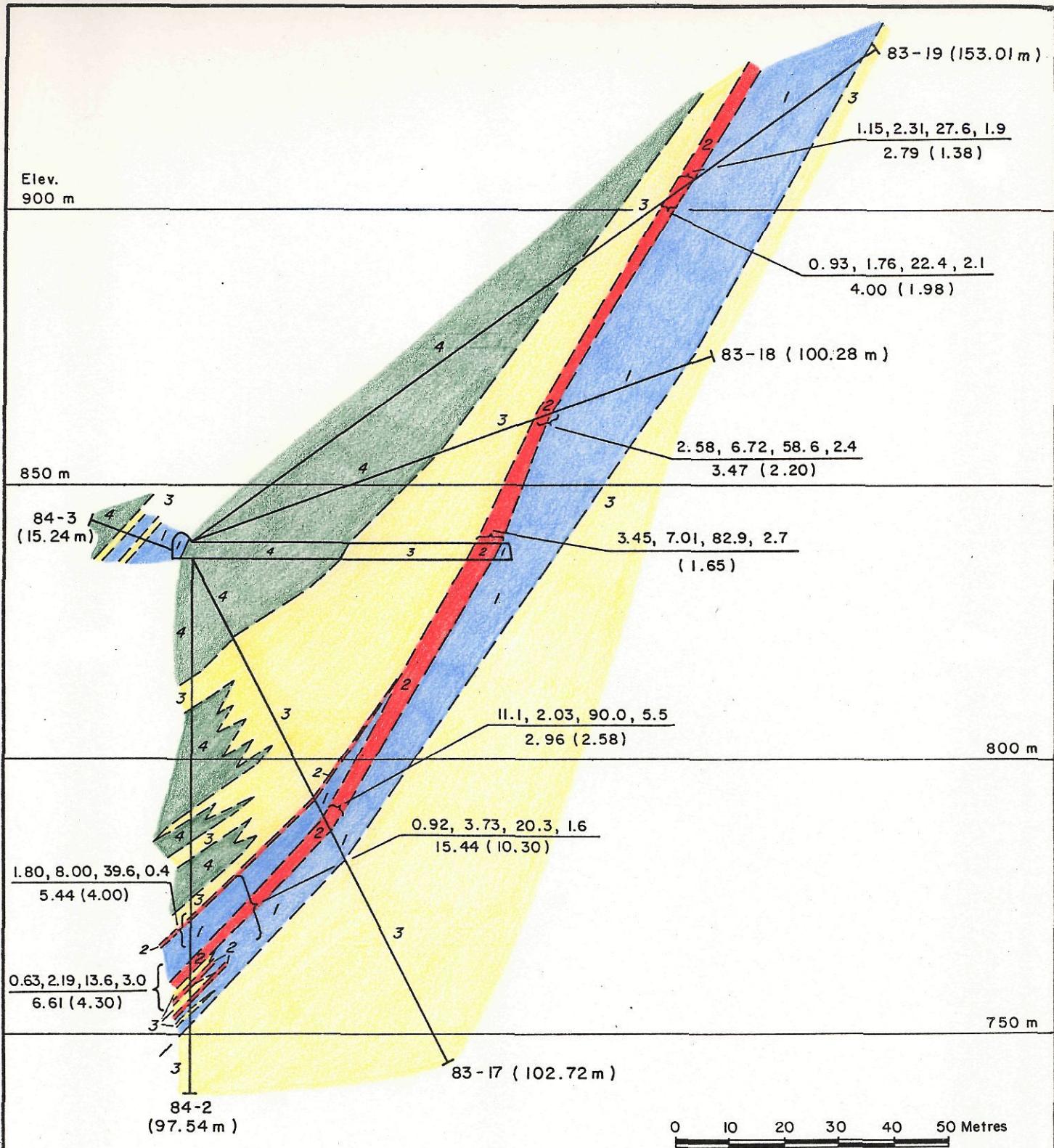
Ore reserve calculations were completed using a zone which is one of 70% sulphides and a 3 metre width above the drift level and 5 metres below the drift level. A specific gravity of 3.5 was used and the Westairs' data from the 830 m drift was ignored as their grades were essentially confirmed by the 1983 chip sampling, but Selco data indicates a true width of less than 1 metre. Additional chip sampling should be done to investigate this discrepancy. Indicated ore reserves from, approximately, 10329E to the end of the present drift and up to 30 metres below surface and down to 200 metres below the drift level, total 2.764 million tonnes (See Plan 23). Jim Wan of Selco, Toronto has recently completed reserve calculations, using 1.6 m mining width and a 3.4 g/t gold equivalent, outlining 4 zones totalling 5,870 tonnes per vertical metre

grading 2.12% lead, 3.91% zinc, 5.70 grams gold, 58.04 g silver and 4.87% arsenic with an average true width of 2.68 m. The author has not yet verified Wan's figures.

4. Diamond Drill Results

Results from the crosscut drilling indicates that the Main Zone is comprised of between 1 and 10 massive sulphide bands which range between 0.6 to 1.93 metres wide (core length). These bands are separated by sericite-quartz to quartz-sericite phyllites, limestones and quartzites which contain appreciable disseminated mineralization. Zonations within the Main Zone cannot be correlated up and down dip. Mineralized sections range up to 13.00 metres, true width and there is a trend of zone thinning, up dip (See Table V). The general trend of increasing grade and/or thickness of the Main Zone with depth was confirmed by the 10500E and 10820E crosscut drilling, but not in the 10670E crosscut drilling. Generally, the hangingwall and footwall stratigraphy consists of intercalated sericitic quartzites and phyllites and chloritic phyllites (See Plans 5-10).

In the 10500E crosscut, the short hangingwall hole (See Figure 32) exposed interbedded quartzites and limestones which are overlain by chloritic phyllites. The vertical

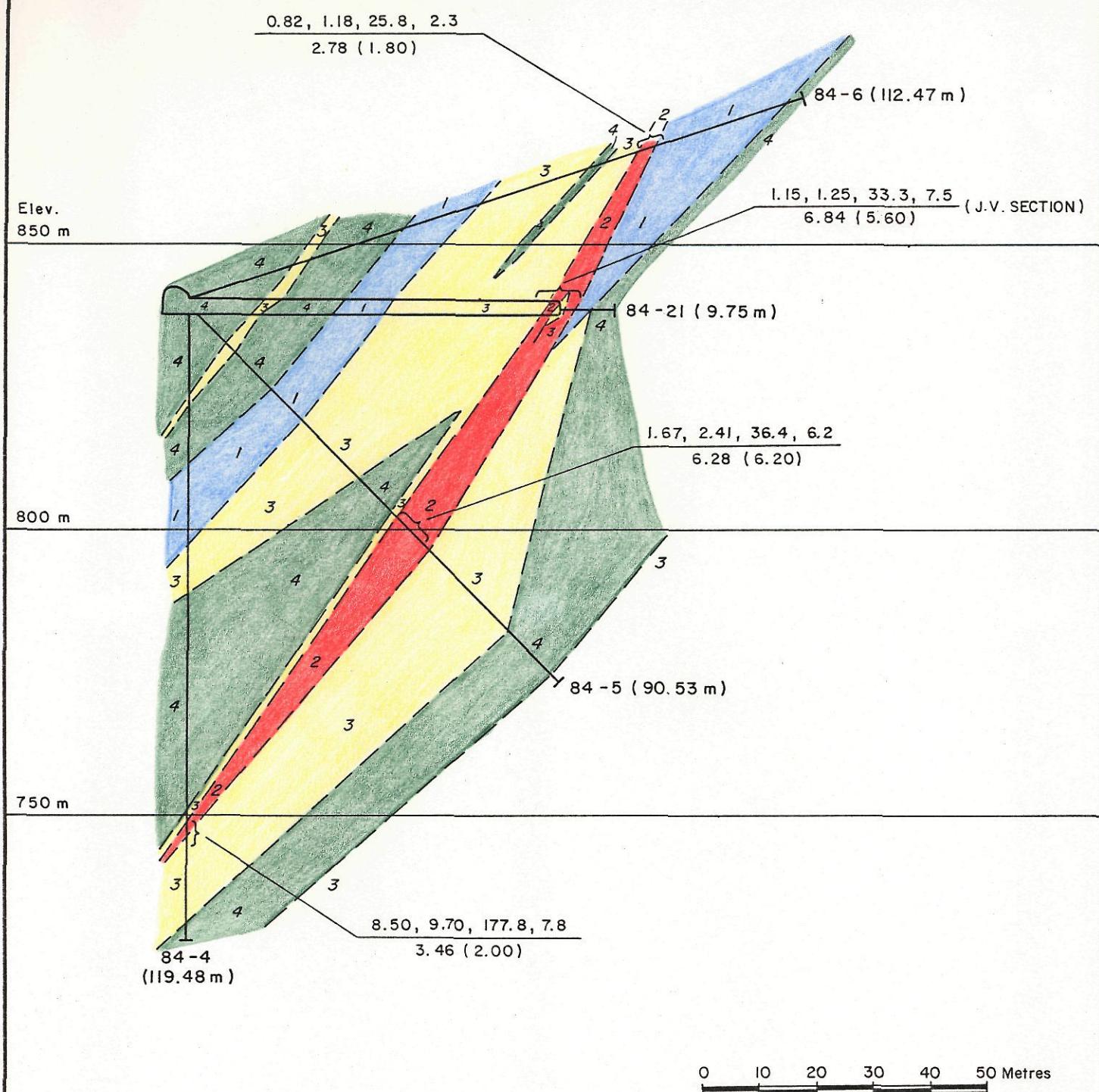


LEGEND

- 4 Chlorite Schists
- 3 Sericitic Quartzite & Chert
- 2 Massive & Disseminated Sulphide
- 1 Limestone

I.11, 2.03, 90.0, 5.5 % Pb, % Zn, g/t Ag, g/t Au
2.96 (2.58) core length (true width), metres

BP		SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT			
SECTION - 500 CROSSCUT			
LOOKING SOUTH			
SCALE	1:1000	DRAWN BY:	R. PEGG
DATE	NOV. 1984	DRAFTED BY:	C.H.K.
M.L.S.	82M/8E	PROJ.	10120
FIG. 32			
REPORT BPVR 84-53			

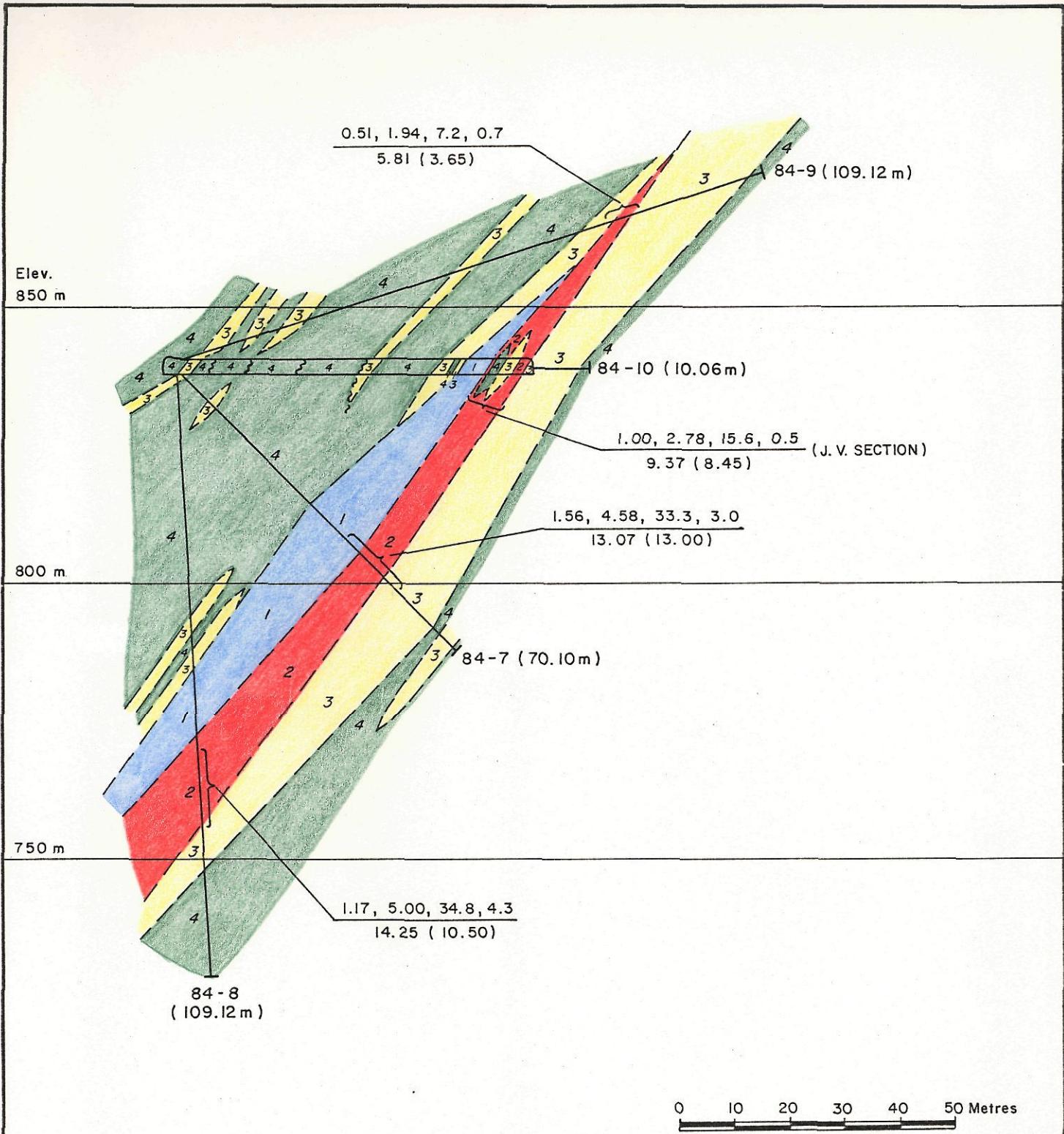


LEGEND

- 4 Chlorite Schists
- 3 Sericitic Quartzite & Chert
- 2 Massive & Disseminated Sulphide
- 1 Limestone

8.50, 9.70, 177.8, 7.8 % Pb, % Zn, g/t Ag, g/t Au
3.46 (2.00) core length (true width), metres

SELCO DIVISION - BP RESOURCES CANADA LIMITED	
J & L PROJECT	
SECTION - 670 CROSSCUT	
LOOKING SOUTH	
SCALE 1:1000	DRAWN BY: R. PEGG
DATE NOV., 1984	DRAFTED BY: C. H. K.
N.T.S. 82 M/8E	PROJ. 10120
REPORT BPVR 84-53	



LEGEND

- 4 Chlorite Schists
- 3 Sericitic Quartzite & Chert
- 2 Massive & Disseminated Sulphide
- 1 Limestone

1.17, 5.00, 34.8, 4.3 % Pb, % Zn, g/t Ag, g/t Au
14.25 (10.50) core length (true width), metres



SELCO DIVISION -
BP RESOURCES CANADA LIMITED

J & L PROJECT
SECTION - 820 CROSSCUT
LOOKING SOUTH

SCALE 1:1000	DRAWN BY R PEGG	FIG. 34
DATE NOV, 1984	DRAFTED BY C H K	
M.T.S. 82M/8E	PROJ. 10120	REPORT SPVR 84-53

hole indicated that the Main Zone consists of 2 sub-zones, the upper of which is zinc rich and gold poor while the bottom zone is lower in base metals, but contains an elevated gold content. Limestones and phyllites separate the individual massive sulphide sections. The weighted average grade of the Main Zone which was intersected by the drilling from the 10500E crosscut is 1.48% Pb, 3.50% Zn, 2.772% As, 47.3 g/t Ag and 3.2 g/t Au over a true width of 2.54 metres.

In the 10670E crosscut, the drilling indicated that the Main Zone is widest in the -45° drill hole (See Figure 33) and thins up and down dip. The up hole also showed a lower grade ore section. The limestone in the hangingwall remains fairly constant in thickness, but the limestone in the immediate footwall pinches out below the drift level and thickens up dip. The weighted average grade of the Main Zone which was intersected by the drilling from the 10670E crosscut is 2.26% Pb, 2.79% Zn, 4.255% As, 52.2 g/t Ag and 6.4 g/t Au over a true width of 3.90 metres.

The drilling from the 10820E crosscut indicated that the zone thickens down dip, but the thickness and grade of the economic sections remains fairly constant below the drift level. This drill section displays relatively low gold values with an increase in gold content down dip (See

N. W.

S. E.

Elev.
900m

83-19 O 1.15, 2.31, 27.6, 1.9
2.79 (1.38)

83-18 O 2.58, 6.72, 58.6, 2.4
3.47 (2.20)

850 m

84-6 O 0.82, 1.18, 25.8, 2.3
2.78 (1.80)

84-9 O 0.51, 1.94, 7.2, 0.7
5.81 (3.65)

800 m

83-17 O 1.11, 2.03, 90.0, 5.5
2.96 (2.58)

84-5 O 1.67, 2.41, 36.4, 6.2
6.28 (6.20)

84-7 O 1.56, 4.58, 33.3, 3.0
13.07 (13.00)

84-2 O 0.92, 3.73, 20.3, 1.6
15.44 (10.30)

84-4 O 8.50, 9.70, 177.8, 7.8
3.46 (2.00)

84-8 O 1.17, 5.00, 34.8, 4.3
14.25 (10.50)

LEGEND

84-2 O

DIAMOND DRILL HOLE
MINERALIZED INTERSECTION

0.92, 3.73, 20.3, 1.6
15.44 (10.30)

% Pb, % Zn, g/t Ag, g/t Au
core length (true width), metres

0 10 20 30 40 50 100 metres



SELCO DIVISION -
BP RESOURCES CANADA LIMITED

J & L PROJECT LONGITUDINAL SECTION 830 m LEVEL ADIT

SCALE	DRAWN BY:	R. PEGG	FIG. 35
DATE DEC., 1984	DRAFTED BY:	C.H.K.	
N.T.S. 82 M/8E	PROJ.	10120	REPORT BPVR 84-53

TABLE VI**
DIAMOND DRILL RESULTS - SUMMARY

Hole Number	Sample From	Inter.(m)	Weighted Average grade						Est. True Width(m)
			% Pb	% Zn	% As	g/t Ag	g/t Au		
84-2	64.56	70.00	1.80	8.00	0.073	39.6	0.4	4.0	
	73.39	80.91	0.63	2.19	2.300	13.6	3.0	4.30	
	64.56	80.91	0.92	3.73	1.159	20.3	1.6	10.30	
84-4	89.08	92.54	8.50	9.70	2.445	177.8	7.8	2.00	
84-5	49.79	56.07	1.67	2.41	3.829	36.4	6.2	6.20	
	49.79	57.99	1.36	1.88	3.363	32.9	5.0	8.00	
	49.79	59.42	1.22	1.65	2.947	29.7	4.5	9.40	
84-6	83.06	85.84	0.82	1.18	2.071	25.8	2.3	1.80	
84-7	41.88	51.68	2.06	5.99	3.963	42.4	4.0	9.80	
	41.88	54.95	1.56	4.58	3.035	33.3	3.0	13.00	
84-8	68.12	82.37	1.17	5.00	3.937	34.8	4.3	10.50	
84-9	79.38	85.20	0.51	1.94	0.229	7.2	0.7	3.65	

**Please note that the weighted average grades were calculated over the same massive sulphide sections, but utilized various widths.

Figure 34). There is no footwall limestone and the hangingwall limestone pinches out above the drift level and is thinning out in the -90° hole. The weighted average grade of the Main Zone which was intersected by the drilling from the 10820E crosscut is 1.42% Pb, 4.72% Zn, 2.819% As, 31.5 g/t Ag and 3.0 g/t Au over a true width of 7.39 metres.

Thus, there is a general trend of increasing zone thickness along strike to the east, in the 1984 crosscut drilling (See Figure 35).

GEOLOGICAL SETTING

Recent studies by structural geologists has fueled the on-going discussion of whether the J & L was originally a vein or sedex type deposit. McClay argues that the deposit is sheet-like and occupies a shear zone which the authors do not dispute. McClay goes on to argue that, due to the lack of primary or syndepositional textures and the presence of calcite and/or quartz fragments in the massive sulphides, the J & L was originally a vein type deposit. He also states that "Even if the sulphides were deposited syngenetically with the host sediments, the intense polyphase deformation would have destroyed any evidence of this origin and has remobilized the sulphides into their present shear zone configuration. The authors believe that

the presence of massive and/or disseminated sulphides at or near the present top of the Hamill limestone unit on Goat Mountain is evidence of primary stratigraphic control. Primary structures, except for gross bedding features, are not expected in a starved basin environment and would easily be destroyed by initial definition. The lack of arsenical sulphides extending into the Mohican metasediments also tends to throw question on the vein origin proposal. The calcite and/or quartz fragments in the sulphides maybe remnants of the country rocks and/or later stage vein events. The presence of the arsenopyrite should not be used as a primary reason for the vein premise as many sedex deposits (e.g. Sullivan) contain arsenic in appreciable, albeit relatively minor, quantities. It should be noted that lithogeochemical work by Gonzalo (1984) on the portal section of the 830 m drift indicates a cyclic nature to the sedimentation. It also indicates that the Main and West Zones are hosted in a stratigraphy that is enriched in carbonate, MgO, Al₂O₃ and K₂O. Gonzalo defined two argillaceous and two quartzitic cycles within the portal section and indicated that the sulphides are found at the top of the argillaceous cycle. XRF results from the underground samples recently submitted to MESA may shed more light on this aspect of the study. In any event, the authors believe that there is a structural overprint on the J & L sulphide system which has been controlled by the stratigraphy.

Computer work by Haworth supposedly indicated that the arsenic-gold and lead-zinc-silver are found in 2 unrelated populations. This and the relative enrichment of base metals with respect to the limestone host/footwall appears to indicate that there are 2 mineralizing events. The author believes that there is a possibility of a lead-zinc-silver system overprinting an arsenic-gold system, but this may just be due to post mineralizing remobilization and segregation.

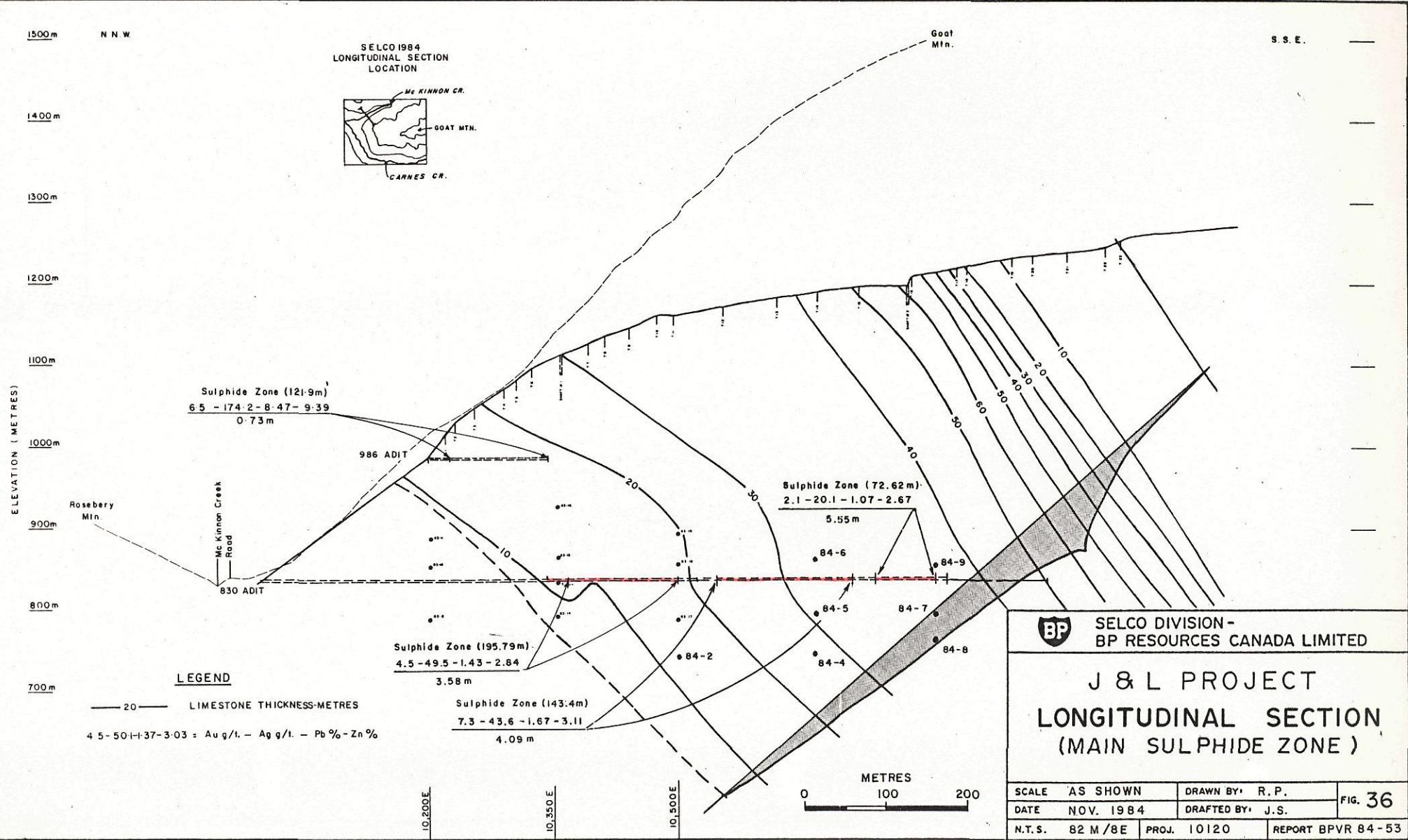
WORK IN PROGRESS

Forty-seven lithologic chip samples from 11 sections in the 830 metre drift and 361 samples from the 10350E and 10500E crosscuts have been submitted to MESA Labs in England for X.R.F. analysis and results are pending. This data could be important in delineating possible indicators of similar deposits in this and other areas and could possibly give us an indication of the origin of the deposit.

The latest metallurgical sample which was submitted to Lakefield Research in the fall of 1984 is still being tested in an attempt to improve the results. Another sample which is comprised of grab samples of the various sulphide types has also been sent to Sunbury in England for further evaluation.

Todd McKinlay, a MSc. geology student at the University of Western Ontario, has been retained to do a study of the J & L deposit. His thesis will include a description of the deposit and its host rocks, interpretation of the genetic significance of this description and recommendations for future exploration. The nature of the occurrence and the significance of the arsenic will be studied. Data collection has and will include surface and underground geological mapping, thin section, fluid inclusion and lead isotope studies, electron probe analyses of individual mineral species and chemical analyses of whole rocks. A comparison of the J & L to other polymetallic sulphide deposits will also be attempted. McKinlay's work, which began during the 1983 field season, will hopefully be finished this year, but progress reports have been few and far between.

Environmental monitoring of the waters draining the J & L deposit is continuing, sporadically, throughout the year. This study is still in progress in order to provide essential background information in the event that mine development at the J & L is considered.



CONCLUSIONS AND RECOMMENDATIONS

It appears that the J & L's geologic environment is favourable to host an economic, arsenical massive sulphide deposit.

The 1984 surface program on Roseberry Mountain has successfully extended the North Zone for 1150 metres. This mineralization appears to be the strike extension of the Main Zone found on Goat Mountain and is still "open" to the northwest. The arsenical mineralization has now been traced for 3.34 km on surface and only 810 metres of this has been investigated underground.

Tight geologically controlled drifting and the subsequent flat drill hole program has proved quite successful in following the Main Zone. The 1984 underground development and drilling has indicated a general increase in the grade and thickness of the Main Zone. Two new mineralized zones have been outlined at the 830 m level, with the second one still "open" along strike. Drilling has indicated that the mineralization is thinning up dip, but is quite significant down dip. Rough tonnage calculations were completed on the Selco data and the indicated ore reserves are 2.764 million tonnes.

The structural study has confirmed the complex structural history of the J & L deposit. Since the idea that the J & L is a

sedex-type deposit has not been disproved, we should continue to keep this concept in mind and hope for a thickening of the sulphides along strike. It also appears that the detailed structural analysis should be continued, if additional development is approved.

A proposal to extend the 830 metre drift for 480 metres has been submitted and this could increase the indicated ore reserves to, approximately, 5.748 million tonnes. If this program is approved, then it should be noted that the Westairs' part of the drift will have to be slashed and re-tracked in order to facilitate the larger equipment required for the drift extension.

Additional surface mapping is recommended on Roseberry Mountain in order to determine the extent and potential of the North Zone. Assessment work is also needed on the Burke claims. This work will require helicopter support.

The most obvious recommendation is for further and intense metallurgical studies in order to test the viability of the extraction of economic minerals from this unusual deposit.

Respectfully submitted,



Rex Pegg, BASc, P.Eng



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PLATE I: View looking south from North Zone (NS-1). Note compressor trailer on road at centre. On right hand side, moving right there is the old camp, the Selco camp and the Rocbore camp.



PLATE II: Geologist prospecting in North Zone area, Roseberry Mountain.



PLATE III: Miner setting charges in 830 m drift face. Note dark coloured massive sulphide band on left wall and mucking machine in foreground.



PLATE IV: Mucking machine in 10820E crosscut dumping waste material into ore car for removal from the underground workings.



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PLATE V: Selco geologists working around Connor's electrical drill in the 10820E crosscut.



PLATE VI: Main Zone mineralization in back of 830 m drift.
Note banded ore on left, milled ore in centre and
lenses and bands of quartz-sericite phyllite
within.



PLATE VII: Coarse-grained, mineralized quartz lens in back of 830 drift. Note sphalerite and galena fracture filling and sericite phyllite lenses within the quartz. Massive sulphides and sericite phyllite on right.

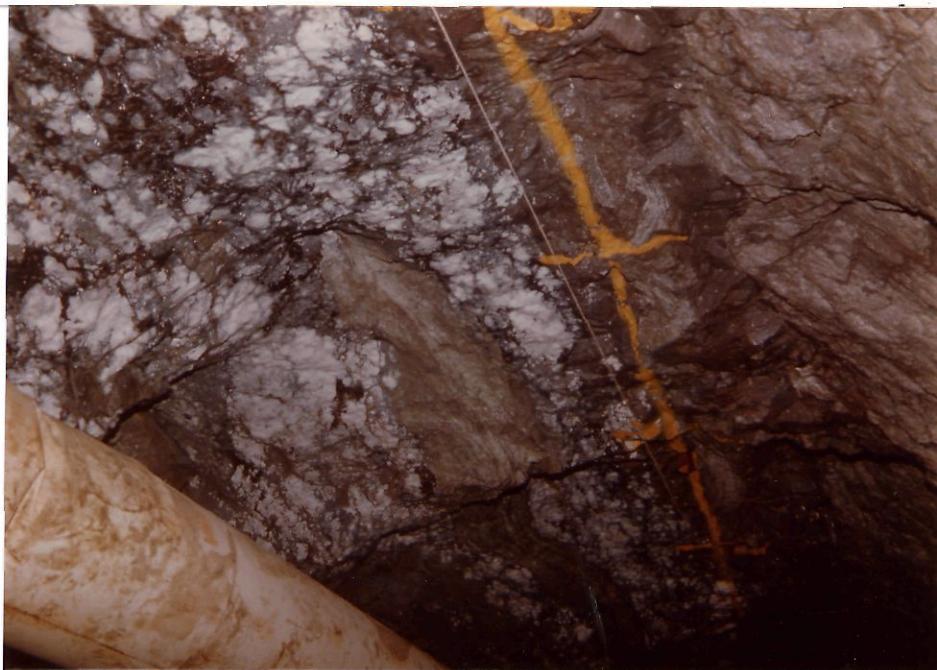


PLATE VIII: Sulphide segregation in highly folded area of 830 m drift.

Note: pyrite and arsenopyrite overlying the light brown narrow sphalerite bands within a grey banded limestone. Quartz-sericite phyllite at the top.



PLATE IX: F4 folding of sphalerite-rich massive sulphides and silicified and mineralized carbonaceous limestone on 830 m drift back at 10825E.
Looking south.



PLATE X: Tight isoclinal folding of brown to honey sphalerite bands and silicified limestone at last drift face in 830 m drift, looking east. This face is at 10830.69E and assayed over 20.5% zinc.

PLATE XI: Sampling pattern of mineralized section near start of 10500E crosscut, looking east. From right to left, intercalated sulphides and phyllite, milled sulphides, banded sulphides, sheared quartz and sphalerite, disseminated phyllite, milled sphalerite and finally quartz-sericite phyllite.



PLATE XII:

Hangingwall phyllites showing injection of massive, milled sulphides and bull quartz tension gashes on north wall of 830 m drift.



APPENDIX I - FIELD PERSONNEL

P. Bartier	Geological Technician	May 5-Oct.14, 1984
L. Garrow	Cook	Jun.18-Oct.20, 1984
T. Garrow	Underground Geologist	May 5-Oct.20, 1984
M. Hislop	Labourer	Sep. 9-Oct.24, 1984
N. Hughes	Geologist	Jun.19-Oct.13, 1984
K. Konkin	Geologist	Jul.19-Aug. 2, 1984
T. McKinlay	Geologist	Jun.19-Sep.14, 1984
J. Michell	Site Engineer	Jul.11-Oct.20, 1984
C. Oke	Geologist	Sep. 9-Oct.24, 1984
R. Pegg	Project Geologist	May 30-Oct.24, 1984
W. Piotrowski	Labourer	Jun.18-Oct.24, 1984
K. Syrja	Labourer	Jul.17-Sep.11, 1984
M. van Wermeskerken	Geology Student	Jun.19-Aug.24, 1984

MAIN ZONE - DRIFT APPENDIX II
FACE GEOLOGY AND GEOCHEMISTRY

J & L LEGEND

- [01] Graphite Schist
- [02] Carbonaceous Limestone and Gert
- [03] Limestone
- [04] Sericite (Muscovite), Sericite-Quartz Phyllite
(maybe calcareous)
- [05] Chlorite, Chlorite-Sericite, Chlorite-Quartz Schist
(maybe calcareous)
- [06] Quartzite, Quartz-Chlorite Schist (maybe calcareous)
- [07] Quartz-Sericite Schist (\pm minor chlorite)
- [08] Chert
- [09] Injected Quartz Breccia
- [10] Mylonite (sheared schist)
- [11] Calcite, Barite, Dolomite
- [12] Massive Sulphides
- [13] Quartz
- [14] Porphyry Dyke

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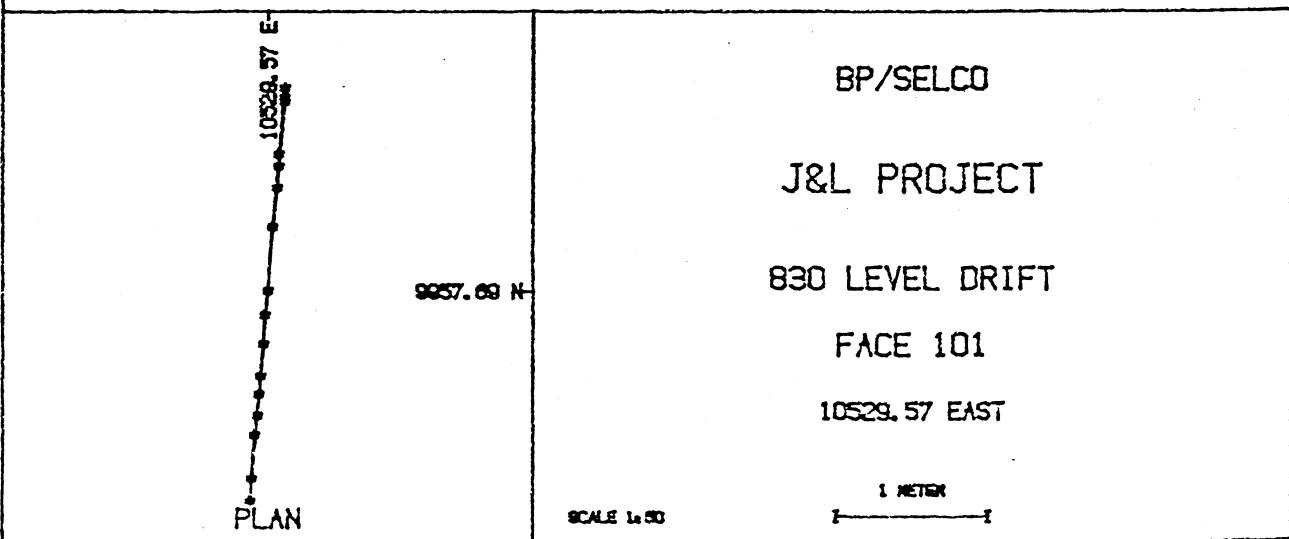
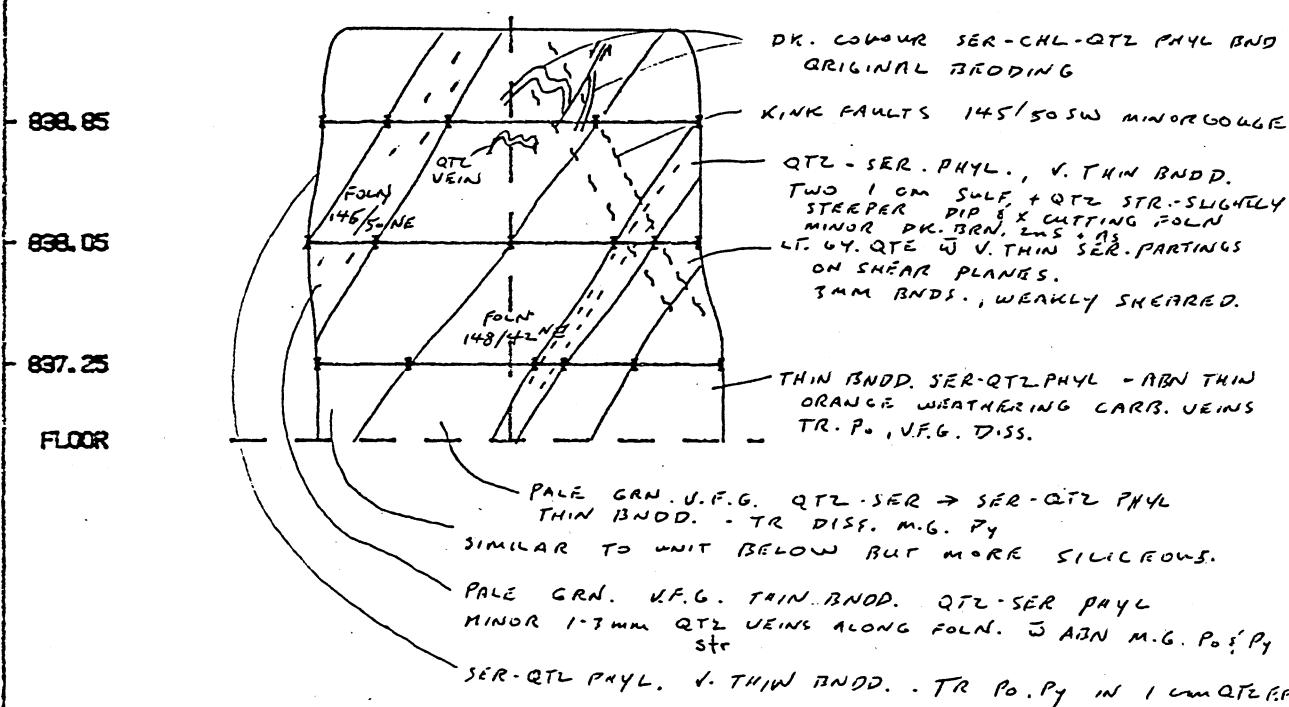
FACE# 101

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43005	.44	.03	.01	.031	5.5	.1
43006	.90	.01	.01	.001	2.3	.4
43007	.68	.01	.01	.001	.3	.1
43008	.27	.03	.34	.582	2.1	1.3
43009	.29	.01	.01	.006	.3	.3

SAMPLE NUMBERS				GEOLOGY CODES					
049001	049002	049003	049004	H	H	=	=		
049005	049006	049007	049008	5	5	3	3		
049010	049011	049012	049013	5	5	3	5	8	
			049014	5	3	5	8	5	

GEOLOGY

ELEVATION



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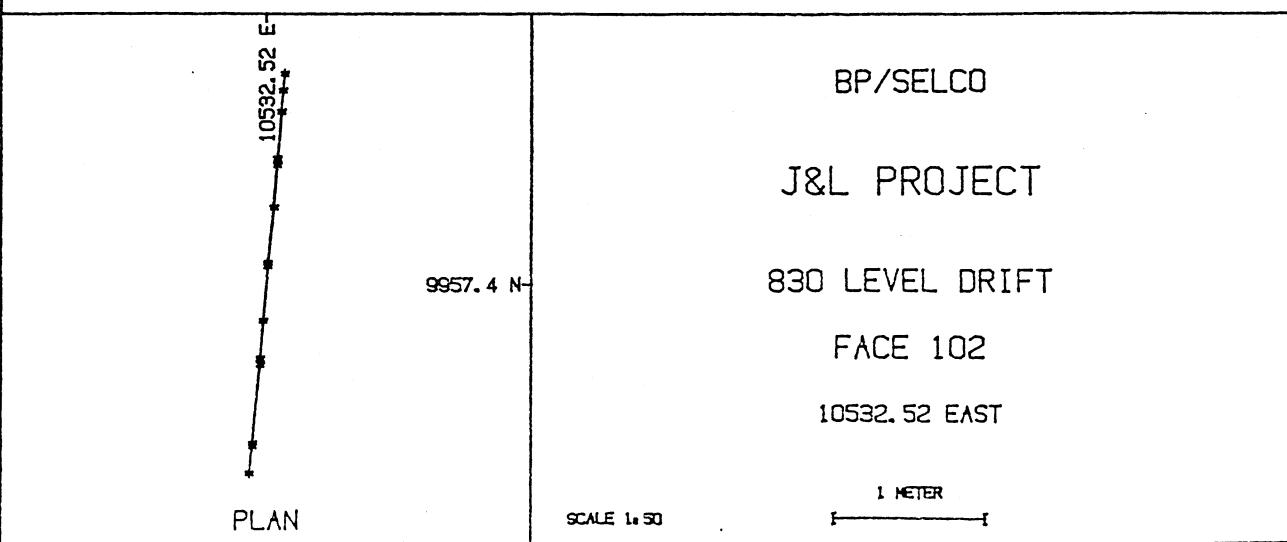
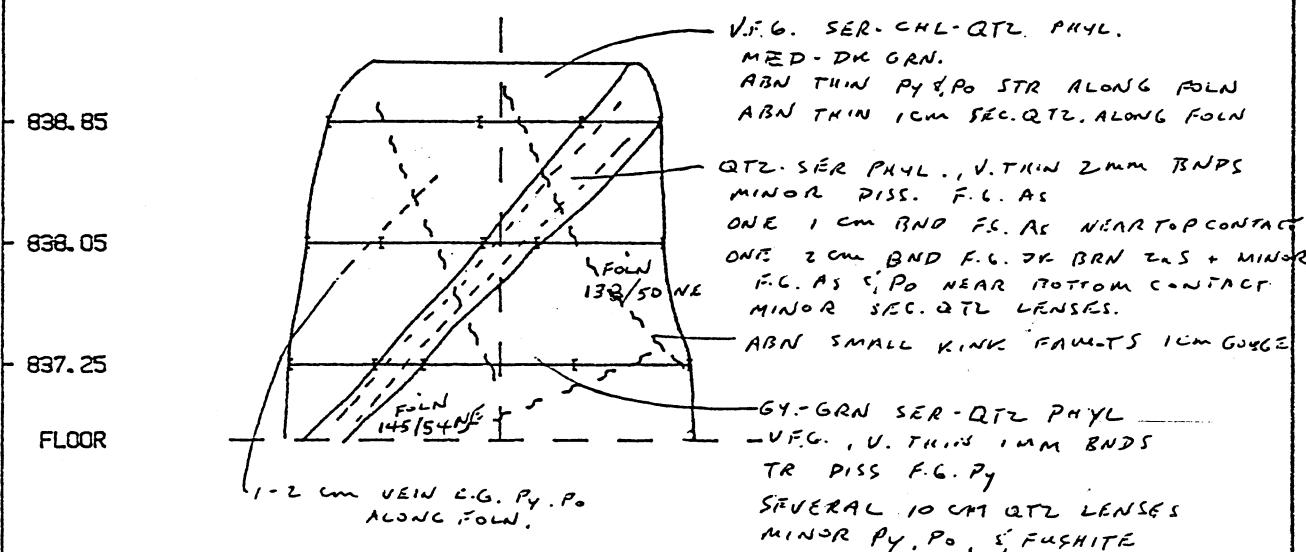
FACE# 102

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43018	.49	.01	.01	.003	3.2	.2
43019	.67	.01	.02	.002	2.8	.1
43020	.36	.03	.31	1.490	2.1	.9
43021	.83	.01	.01	.029	.3	.1

SAMPLE NUMBERS			GEOLOGY CODES		
043015	043016	043017	05 H	05 H	07 H
043018	043019	043020	05 H	05 H	04 H
043022	043023	043024	05 H	07 H	04 H

GEOLOGY

ELEVATION



SELCO DIV. - BP RESOURCES

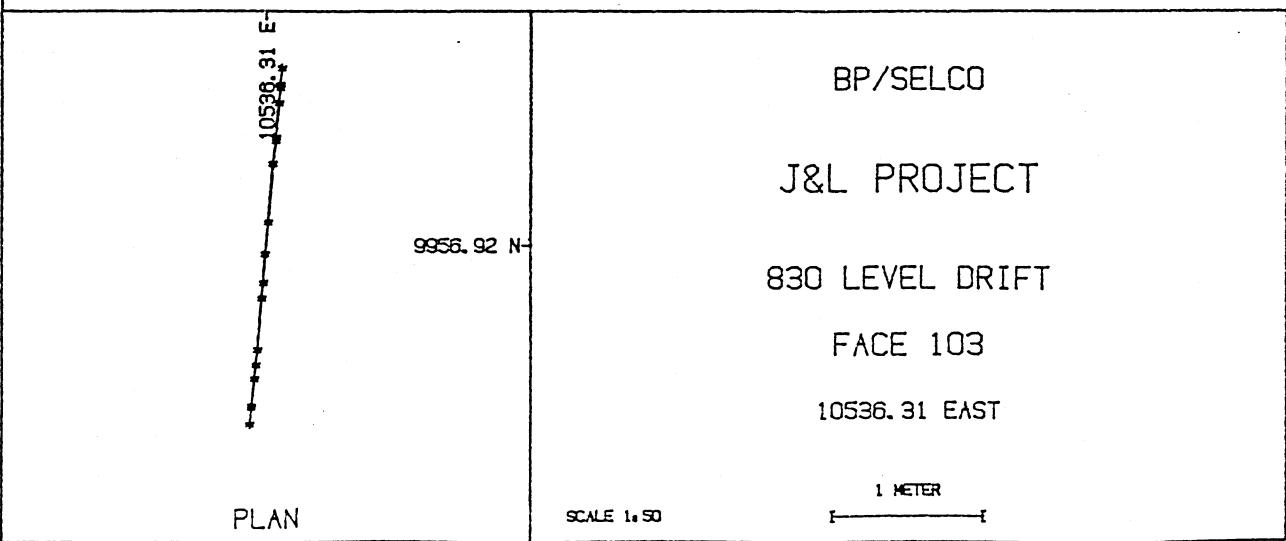
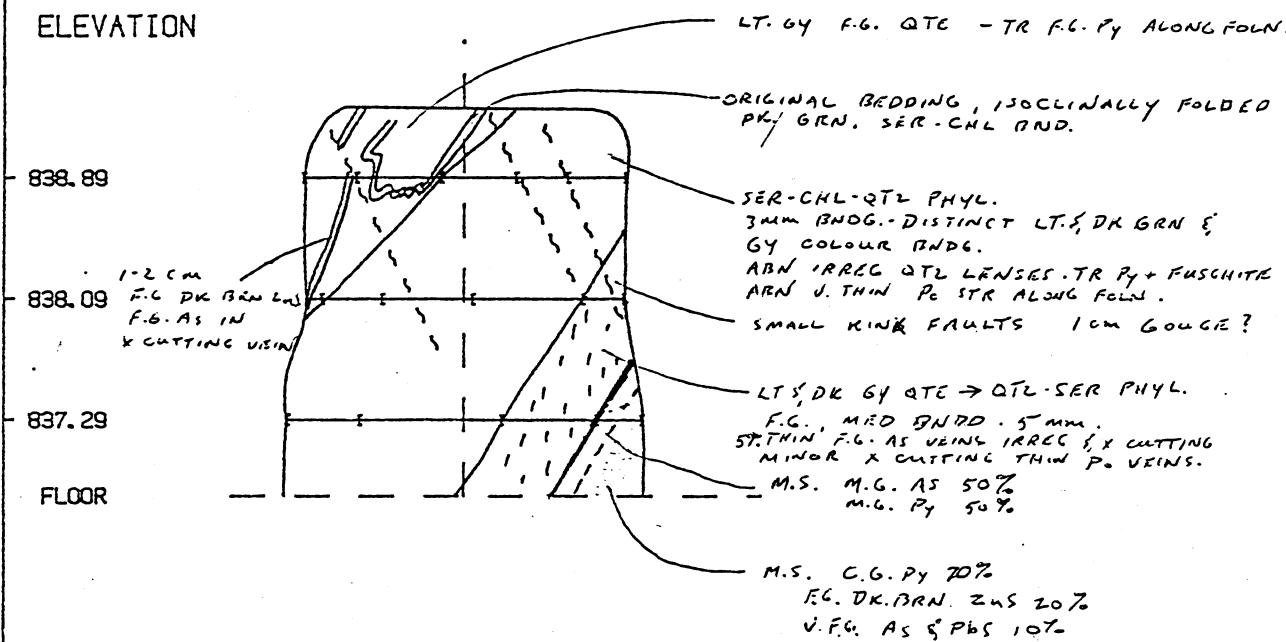
J & L PROJECT, B.C.

FACE# 103

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43031	.10	.01	.02	.083	.7	.1
43032	.40	.01	.01	.004	3.1	.3
43033	.60	.01	.01	.002	1.2	.1
43034	.73	.01	.01	.002	1.7	.1
43035	.27	.02	.01	1.790	15.0	.7

SAMPLE NUMBERS					GEOLOGY CODES				
043026					08	H			
043027	043029	043028	043029	043030	08	H	05	05	05
043031	043032	043033	043034	043035	08	H	05	05	08
043036					05	H	05	05	08
043037	043038	043039			05	H	05	08	12.0

GEOLOGY



SELCO DIV. - BP RESOURCES

J & L PROJECT, B.C.

FACE# 104

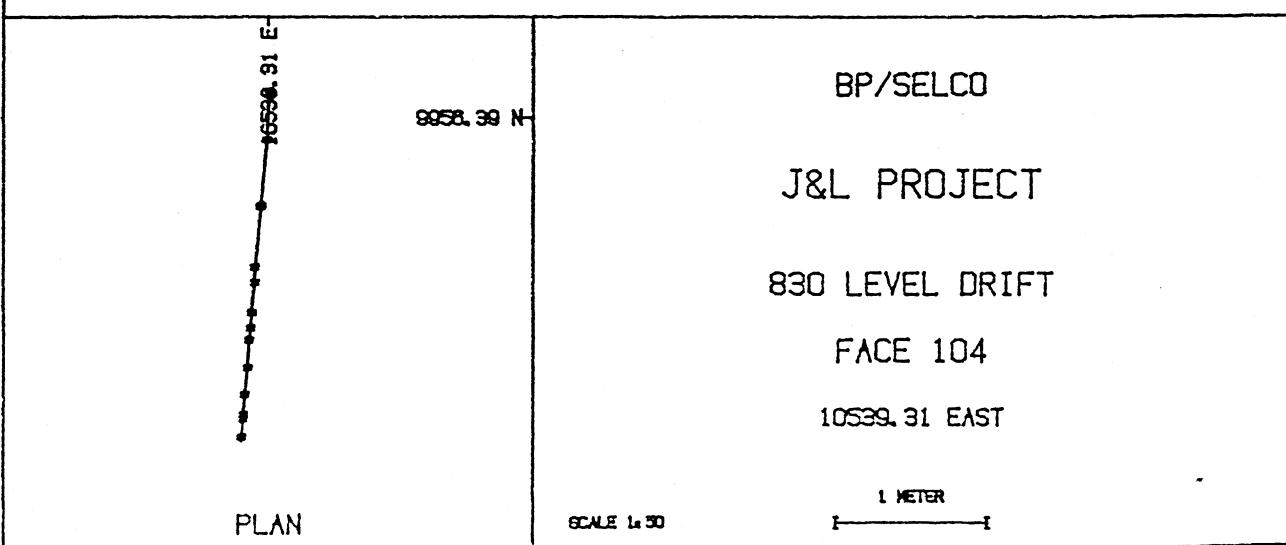
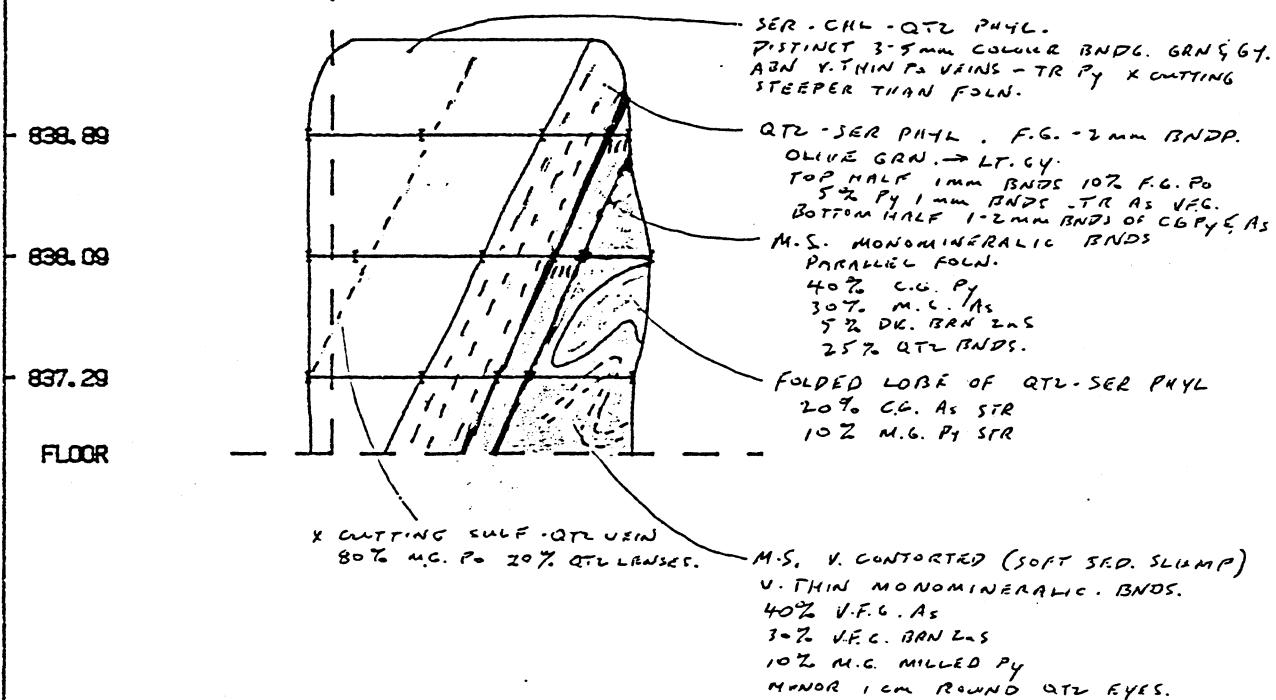
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SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43040	.75	.01	.01	.020	.3	.3
43041	.80	.01	.01	.002	1.0	.2
43042	.44	.02	.01	.380	2.4	.3
43043	.13	6.97	17.20	4.510	211.5	3.8
43044	.31	.06	.13	.031	.3	.2
43045	.84	.02	.02	.005	.9	.3
43046	.48	.01	.01	.214	2.2	.5
43047	.18	.56	.45	4.760	25.3	2.1
43048	.46	9.18	16.80	4.380	259.2	9.5

SAMPLE NUMBERS				GEOLOGY CODES			
49040	049041	049042	049043	H	H	H	12 10
049044	049045	049046	049047	05	05	07	12 0
049048		049049					
049049	049050	049051	049052	05	07	12 0	12 0

GEOLOGY

ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 105

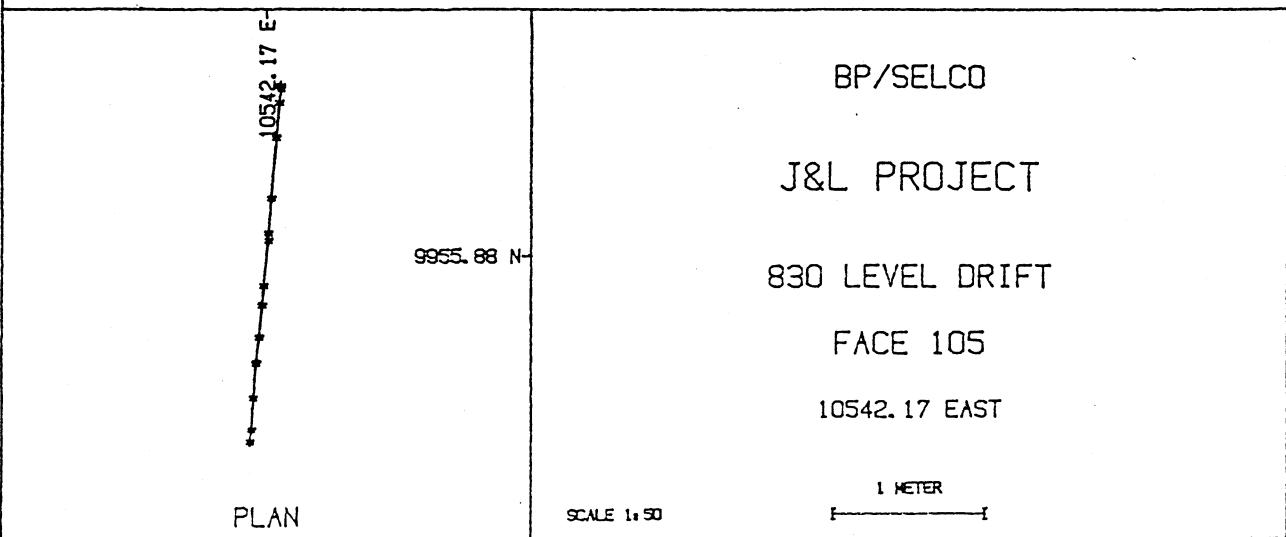
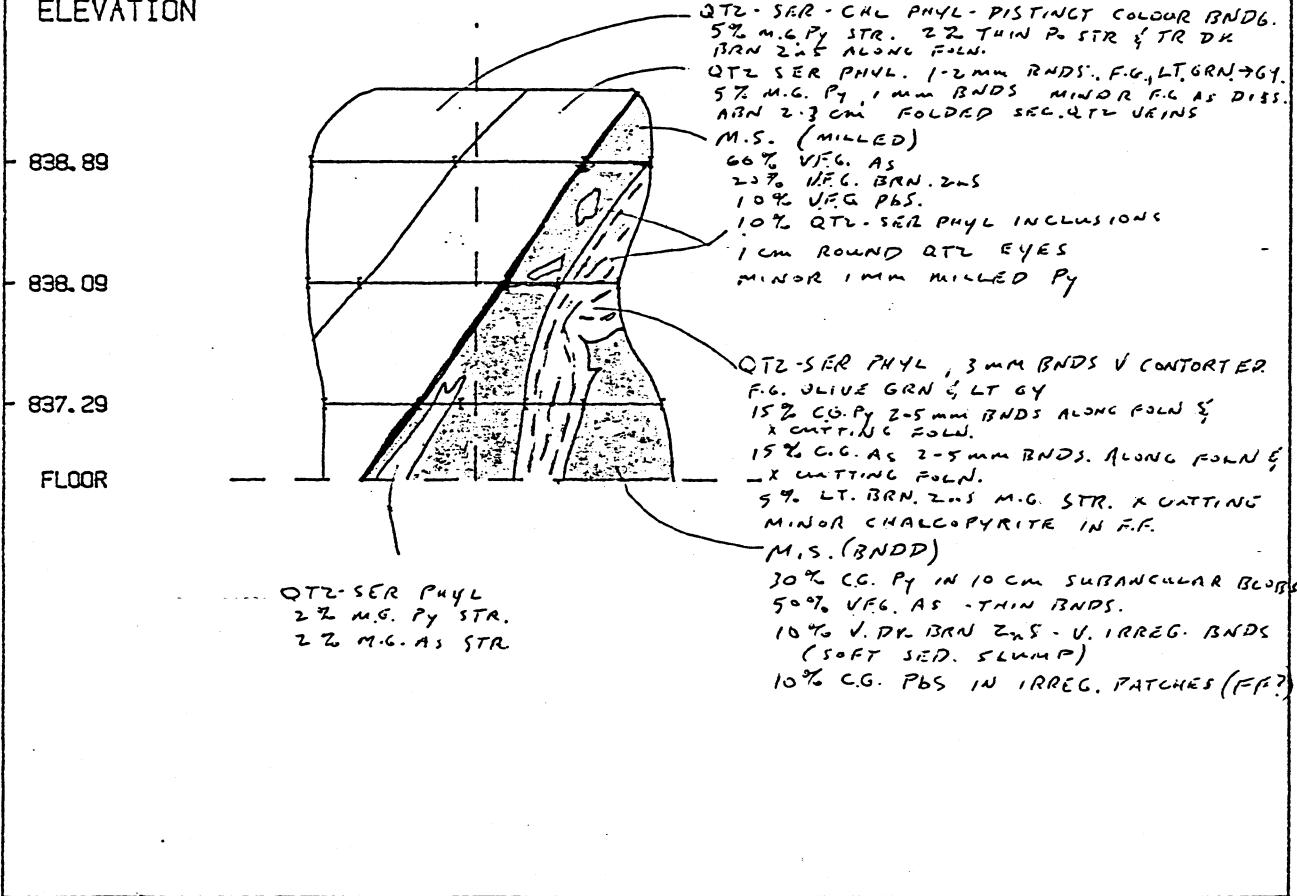
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SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43053	.95	.10	.15	.043	7.5	.1
43054	.86	.06	.03	.360	.9	.3
43055	.44	8.52	22.20	3.240	230.8	10.5
43056	.34	.10	.21	.029	7.2	.3
43057	.98	.04	.08	.039	1.8	.2
43058	.34	11.20	15.80	3.430	278.0	9.3
43059	.40	.43	.17	1.620	10.6	1.0

SAMPLE NUMBERS			GEOLOGY CODES		
043053	043054	043055	H	H	0
043058	043057	043058	07	07	12 0
043060	043061	043062	07	07	12 0
043063	043064		07	07	12 0

GEOLOGY

ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 106

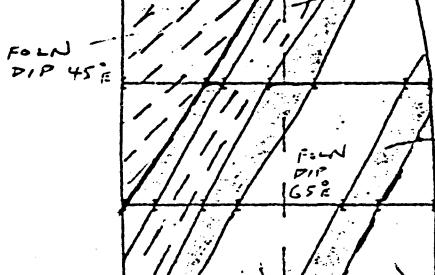
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43065	1.00	.10	.25	1.090	4.8	.7
43066	.11	5.93	22.20	2.810	147.6	6.7
43067	.35	.16	.13	1.570	5.7	1.1
43068	.17	10.10	12.50	10.700	330.3	27.6
43069	.27	.24	.24	2.040	10.9	1.4
43070	.54	.06	.06	.214	3.0	.4
43071	.13	7.90	21.30	3.850	213.5	7.9
43072	.30	.10	.16	1.330	2.9	.5
43073	.30	9.95	20.60	3.930	459.7	16.1
43074	.60	.20	.23	1.010	12.3	.7
43075	.16	.36	.19	.940	20.2	3.8

SAMPLE NUMBERS					GEOLOGY CODES				
043085	043086	043087	043088	043089	H	07	12	0	07
043070	043071	043072	043073	043074	043075	07	12	0	07
043076	043077	043078	043079	043080	043081	12	0	12	0
						07	0	12	0
						07	0	07	02

GEOLOGY

ELEVATION

838.92



838.12

FLOOR

QTL-SER PHYL.
5% UFG AS DISCS. - 10% M.G. PO 1-10 mm BNDS
10% M.G. PY THIN STR ALONG FOLN.
1 CM UND. DK. M.G. ZNS. MINOR PGS F.F. - TR MENEIG.
M.S. MINOR QTL-SER PHYL FRAGS <3 cm
40% VFG. AS
10% VFG. PGS.
5% 1 mm MILLED PY
QTL-SER PHYL - FG. THIN BNDS.
15% CG. AS BNDS 2-4 mm ALONG FOLN.
5% CG. PY BNDS
MINOR F.G. AS DISCS.
20% SEC QTL LENSES ALONG FOLN.

M.S. (BNDS)
TOP HALF MIXED CM BNDS
50% CG AS + 20% CG PY + 5% RED ZNS
+ 25% QTL-SER PHYL.
BOTTOM HALF. ZONED F/W UP AS-PY-ZNS
50% M.G. PY, 35% UFG AS, 15% RED ZNS

BK. ARCL LST F.G. 20% WHITE CALC BNDS
TOP 10 cm WHITE RECRYSTALLIZED LST
MINOR BK. ARCL BNDS 3 mm
MINOR LT. BRN Z-S, TR F.G. PY
F.G. QTL-SER PHYL 4-5 mm MONOMINERALIC
SULF BNDS.
TOP 7% RED F.G. ZNS, MINOR M.G. PY
MIDDLE 5% CG. PY + 5% C.C. AS
BOTTOM 5% CG. PY

M.S. (BNDS) ZONED.
TOP 75% U.D. BRN. M.G. ZNS. MINOR F.G. AS
MIDDLE 60% C.C. PY, 10% F.G. AS BNDS
BOTTOM 60% F.G. AS 10% F.G. LT. RAN ZNS
10% F.G. PY - 2 Z M.G. PLS X CUTTING LENSES

10544.36 E

9953.2 N

BP/SELCO

J&L PROJECT

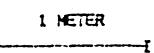
830 LEVEL DRIFT

FACE 106

10544.36 EAST

PLAN

SCALE 1:50



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 107

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43082	.41	.03	.23	.610	5.1	1.1
43083	.13	5.51	23.50	2.810	140.6	6.9
43084	.25	.43	1.88	11.900	12.7	2.4
43085	.23	4.48	16.30	3.030	98.0	8.1
43086	.21	.23	.26	.171	9.9	.3
43087	.79	2.14	13.30	5.330	72.7	8.3
43088	.15	.71	5.74	.345	16.8	1.0
43089	.14	4.84	12.80	3.920	116.9	11.5
43090	.15	1.37	9.39	3.620	34.3	6.8
43091	1.00	3.68	21.30	3.950	63.1	7.0
43092	.83	.81	2.77	1.120	36.0	2.3
43093	.17	2.62	2.53	6.620	41.5	10.8

SAMPLE NUMBERS						GEOLOGY CODES					
043082	043083	043084	043085	043086	043087	07 H	12 0	07 0	12 0	07 0	12 0 F
043088						12 0	12 0	02 F	02	02	F
043089	043090	043091	043092	043093							
043094	043095	043096	043097			0		F			
						07	02	F	02		

GEOLOGY

ELEVATION

838.92

838.12

837.32

FLOOR

043-SER-PHYL.
20% CG. AS + 10mm BNDS
MINOR PY MINOR M.G. PLS FF.
TR MENEG

QTL-SER-PHYL. 1-3 mm BNDS
10% CG. PY 1mm BNDS, MINOR AS PLS
ONE 5mm LENS DK. BRN 2n5. M.G.
M.S. (BNDS) ZONED
TOP 5cm 65% V.G. PY + QTL-SER-PHYL
MIDDLE 80% BK. M.G. 2n5, 20% CG. PY
BOTTOM 90% V.G. AS., TR PY, 2n5

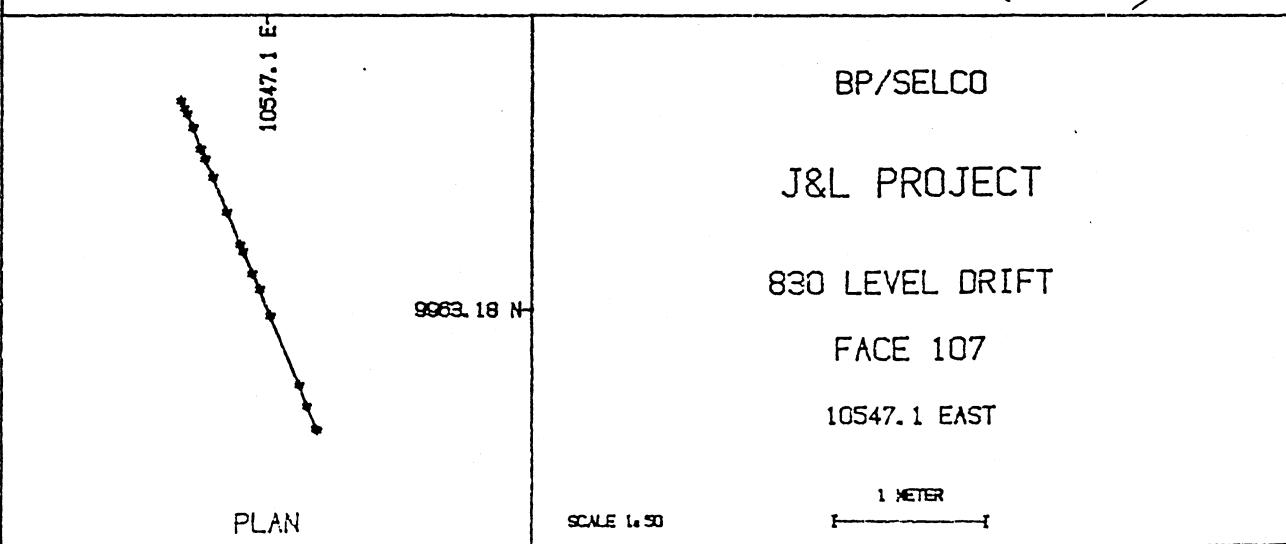
M.S. (BNDS & CONTORTED)
60% CG. PY IN 5-10 cm IRREG BLOBS
30% M.G. BK 2n5
10% V.G. AS
TR MENEG?

QTL-SER-PHYL. F.G. THIN BNDS
10% CG. AS STR + 5% F.G. PY STR
+ MINOR DK. BRN 2n5 STR.
TR CHALOPYRITE X CUTTING F.F.

BK. ARG. LST, TR GR ON FOLN
MINOR 2cm LT GY LST BNDS
10% V.G. CALC SWEAT OUTS 1-3 cm
TR HONEY 2n5 + M.G. PY IN F.F.

BK. F.G. ARG. LST TR GR ON FOLN
40% WHITE CG CALC. IN 2-10 mm
BNDS, V. CONTORTED (SUFT-SED)
ZONED SULFIDES
TOP SEVERAL 3-5 mm BNDS 40% F.G. AS
40% F.G. PY, 20% RED 2n5 BNDS
MIDDLE 5-10 ARD 2n5 2mm BNDS
+ TR PY V. CONTORTED
BOTTOM SEVERAL BNDS 90% V.G. AS
MINOR PY MINOR LT BRN 2n5

M.S. (BNDS & CONTORTED)
GOOD MONOMINERALIC BNDS
30% M.G. PY BNDS
30% RED 2n5 BNDS
20% V.G. AS BNDS
20% QTL-SER-PHYL. BNDS (CONTORTED)

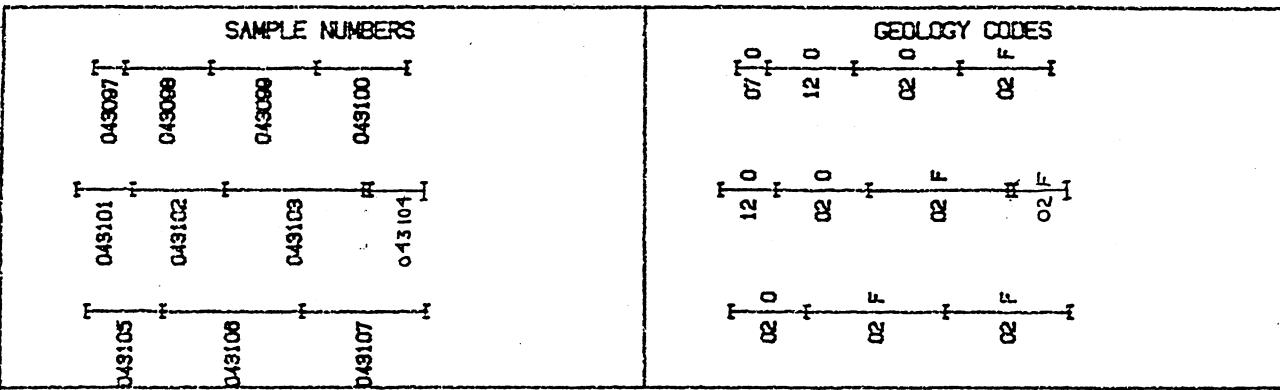


SELCO DIV. - BP RESOURCES J & L PROJECT. B.C.

FACE# 108

* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43097	.20	.30	.85	6.590	3.1	2.3
43098	.57	4.20	17.90	3.620	180.3	9.7
43099	.70	1.51	3.47	1.630	24.3	3.1
43100	.60	.16	1.44	.376	1.7	1.0
43101	.37	12.80	17.40	.987	287.6	10.6
43102	.61	1.45	3.97	1.900	35.0	4.0
43103	.92	.11	.21	.411	.3	1.2
43104	.40	.03	.04	.015	.1	.1



GEOLOGY

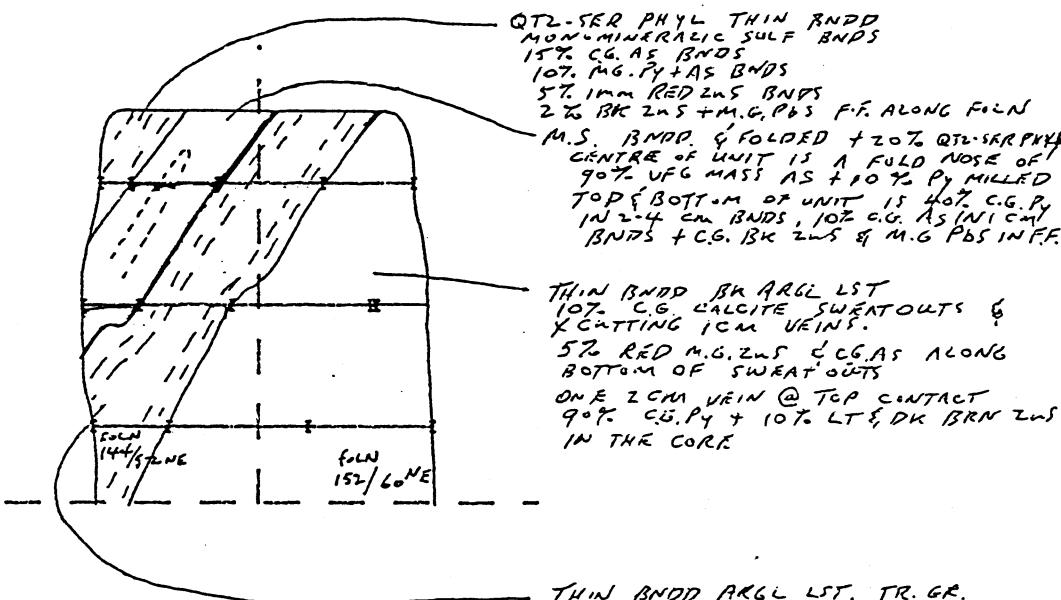
ELEVATION

838.92

838.12

837.92

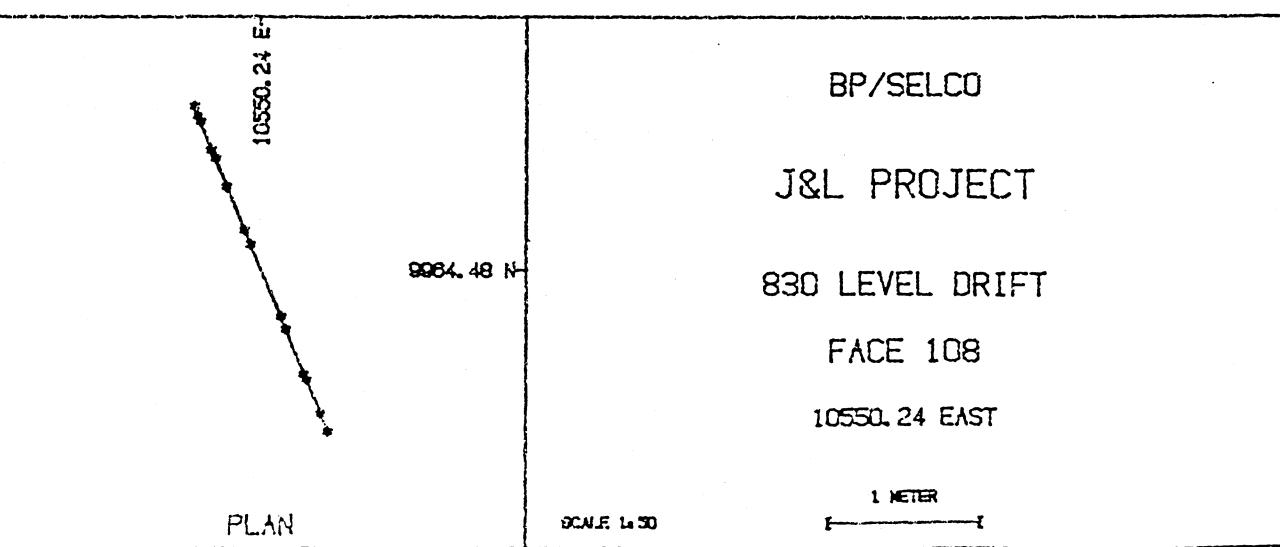
FLOOR



QTL-SER PHYL THIN BNDD
MONOMINERALIC SULF BNDS
15% CG. AS BNDS
10% MG. PY+AS BNDS
5% 1MM RED ZNS BNDS
2 TO BK ZNS + MG. PBS FF ALONG FOLN
M.S. BNDD. Q FOLDED + 20% QTL-SER PHYL
CENTRE OF UNIT IS A FOLD NOSE OF
90% UFG MASS AS + 10% PY MILLED
TOP & BOTTOM OF UNIT IS HOT CG. PY
1 IN 2-4 CM BNDS, 10% CG. AS IN VAGUE
BNDS + CG. BK ZNS & MG. PBS IN FF.

THIN BNDD BR ARGL LST
10% CG. CALCITE SWEATOUTS &
X CUTTING 1CM VEINS.
5% RED MG. ZNS & CG. AS ALONG
BOTTOM OF SWEATOUTS
ONE 2 CM VEIN @ TOP CONTACT
90% CG. PY + 10% LT & DK BRN ZNS
IN THE CORE

THIN BNDD ARGL LST., TR. GR.
2-4MM BNDS
TOP CONTACT .5 CM WHITE
RECRYSTALLIZED LST - 20% F.G. RED ZNS
MINOR AS
BELOW THIS MONOMINERALIC BNDD.
SURFACES F.G. AS 20%, F.G. PY 20%
10% F.G. BRN ZNS.
CENTER 20 CM OF THIN BNDD
ARGL LST - BARREN
BOTTOM (MILLED) VAGUE THIN
BNDS 20% VFG AS RIMMED BY
BRN ZNS



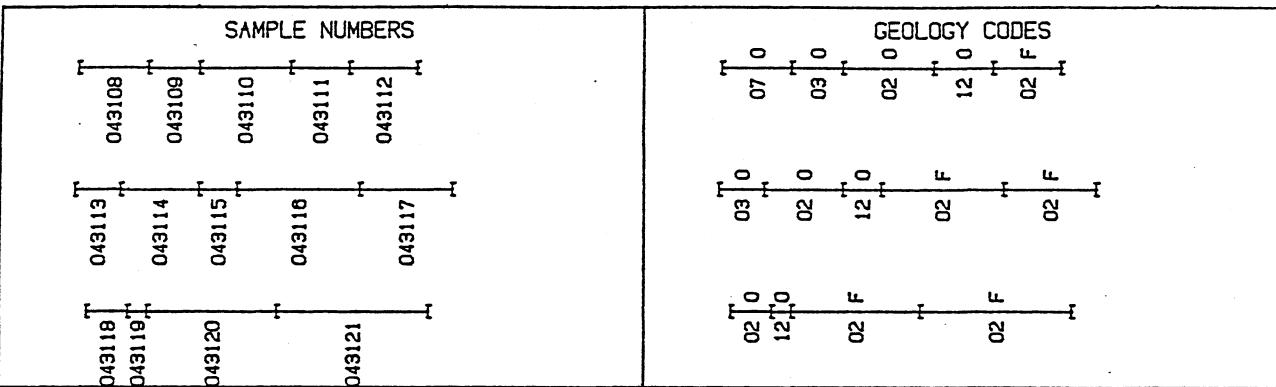
SELCO DIV. - BP RESOURCES

J & L PROJECT. B.C.

FACE# 109

*** *** *** *** *** ***

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (o/mT)	GOLD (o/mT)
43108	.46	.25	.04	2.870	4.5	1.6
43109	.34	1.17	8.52	.128	38.4	.7
43110	.60	.19	.41	.175	10.6	.3
43111	.39	2.42	9.20	3.250	55.2	6.5
43112	.45	.10	1.11	1.000	3.1	2.4
43113	.30	.72	2.96	.319	18.5	.6
43114	.52	.79	3.05	1.350	18.5	4.7
43115	.25	.06	.06	.618	1.4	1.2
43116	.81	.72	1.63	1.420	18.5	2.1
43117	.61	.02	.03	.015	.1	.1



GEOLOGY

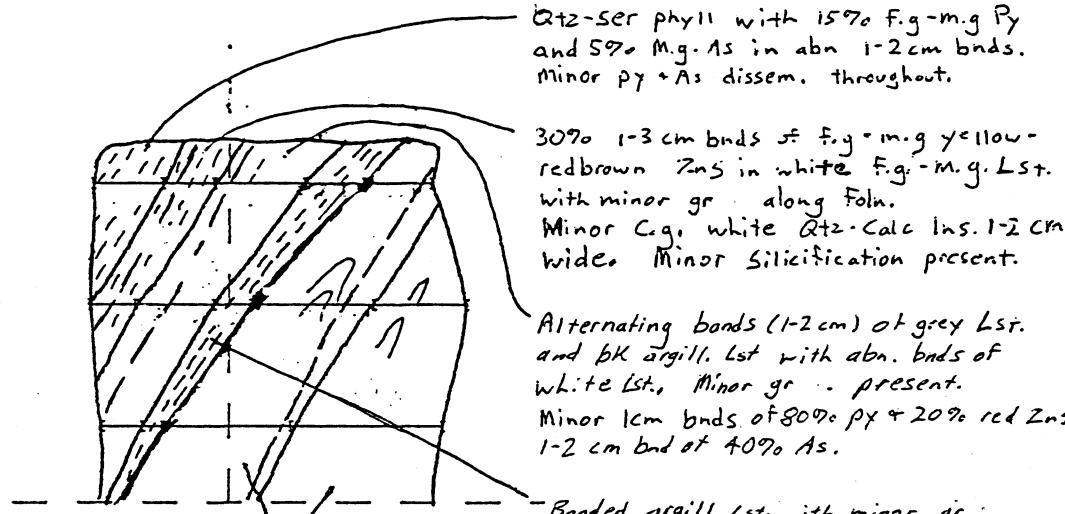
ELEVATION

838.92

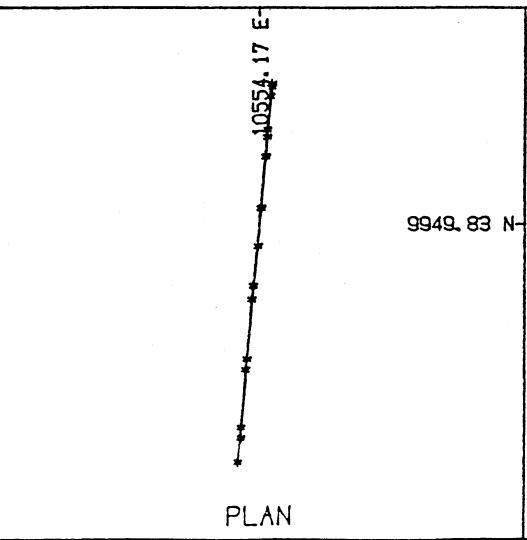
838.12

837.32

FLOOR



Banded argill Lst. with minor gr. 70% 1-2 cm bnds of m.g. As stringers. Minor f.g. Pb + minor mcneng. present. Left 10cm is 60% M.S. 50% f.g. red-brown ZnS, 10% m.g. f.g. Py, 40% argill Lst.



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 109

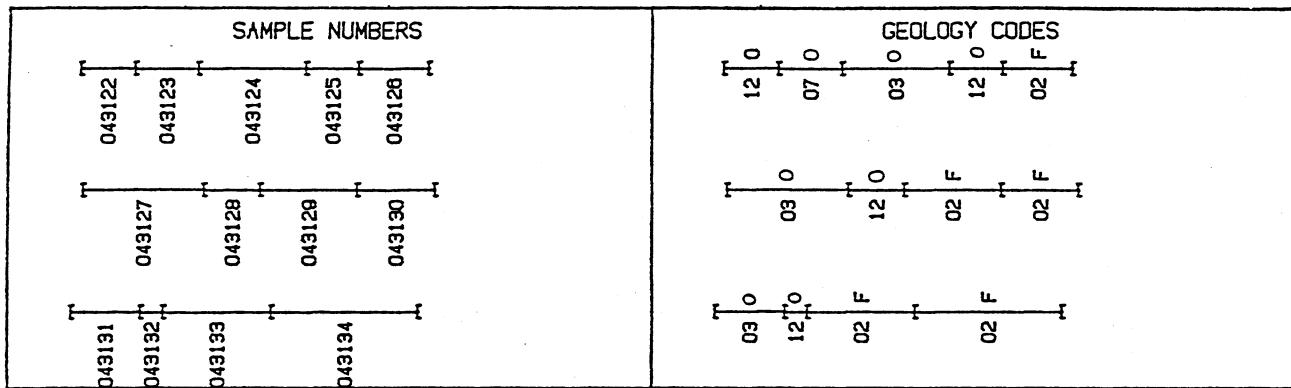
10554.17 EAST

1 METER

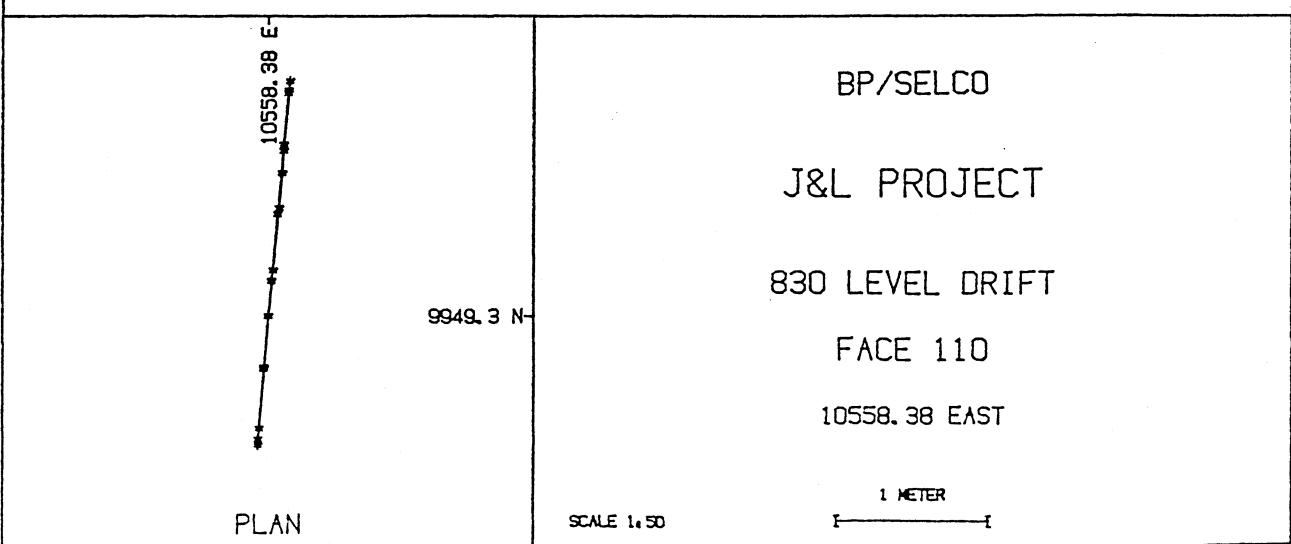
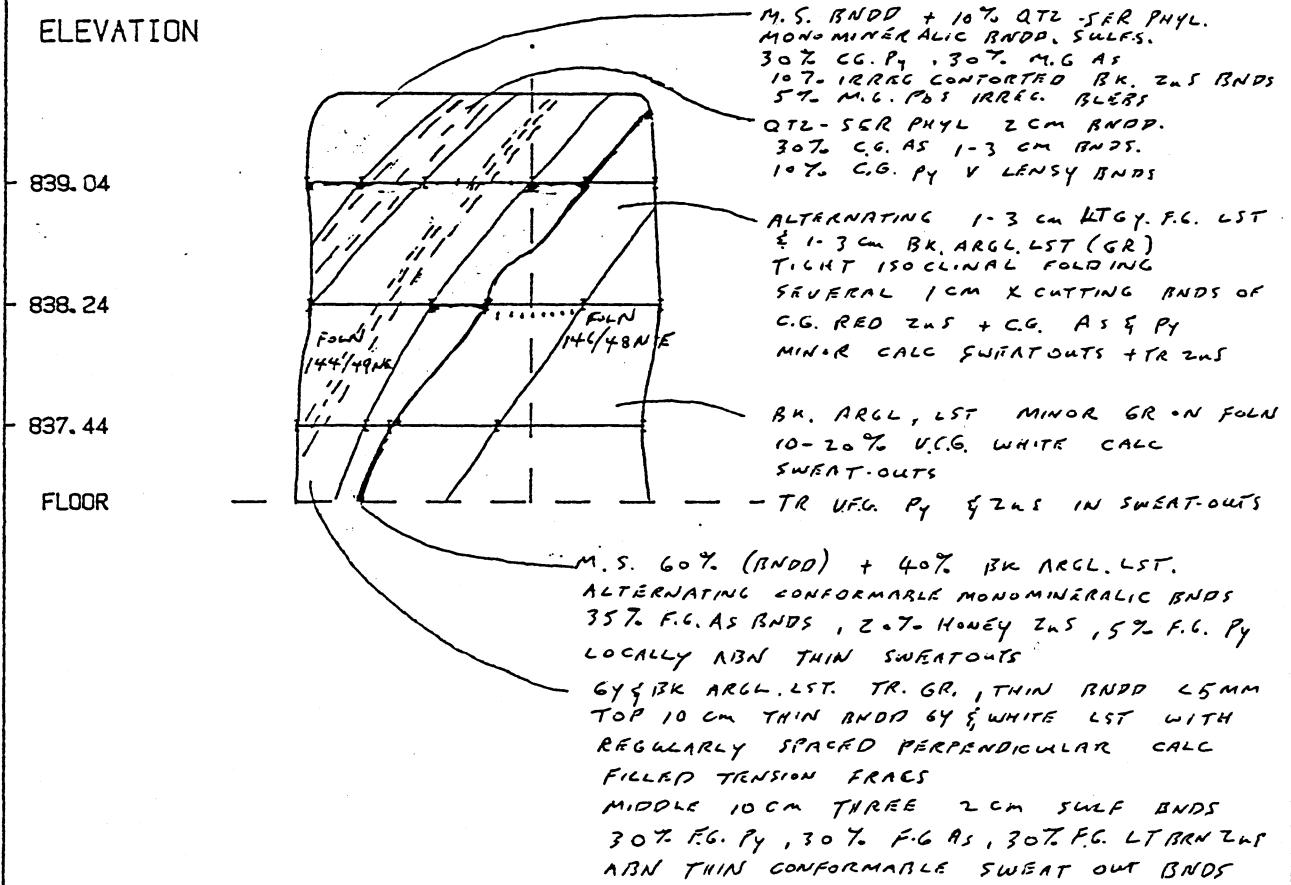
SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 110

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43122	.36	5.40	5.20	10.100	198.5	12.4
43123	.42	1.28	2.35	6.810	52.2	9.5
43124	.71	.65	3.01	1.350	8.8	2.1
43125	.35	1.14	3.05	5.210	36.4	6.8
43126	.46	.16	.20	.210	7.0	.5
43127	.80	.18	.34	.541	9.3	.4
43128	.37	.53	2.18	4.570	20.2	7.1
43129	.64	.30	.31	1.260	8.6	2.5
43130	.51	.02	.07	.024	2.7	.1



GEOLOGY



SELCO DIV. - BP RESOURCES

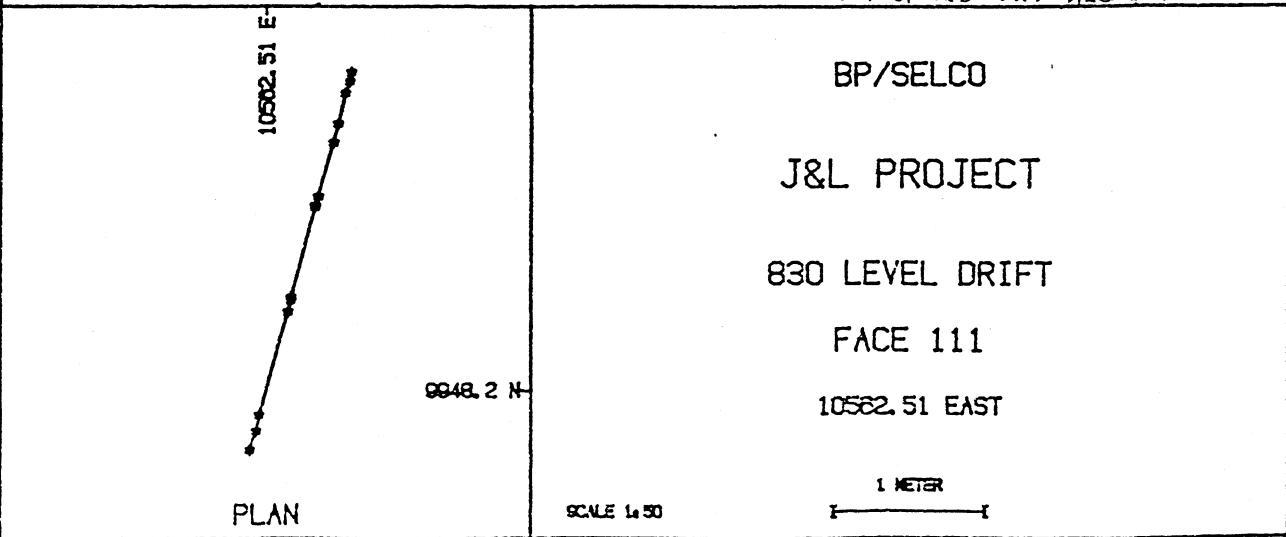
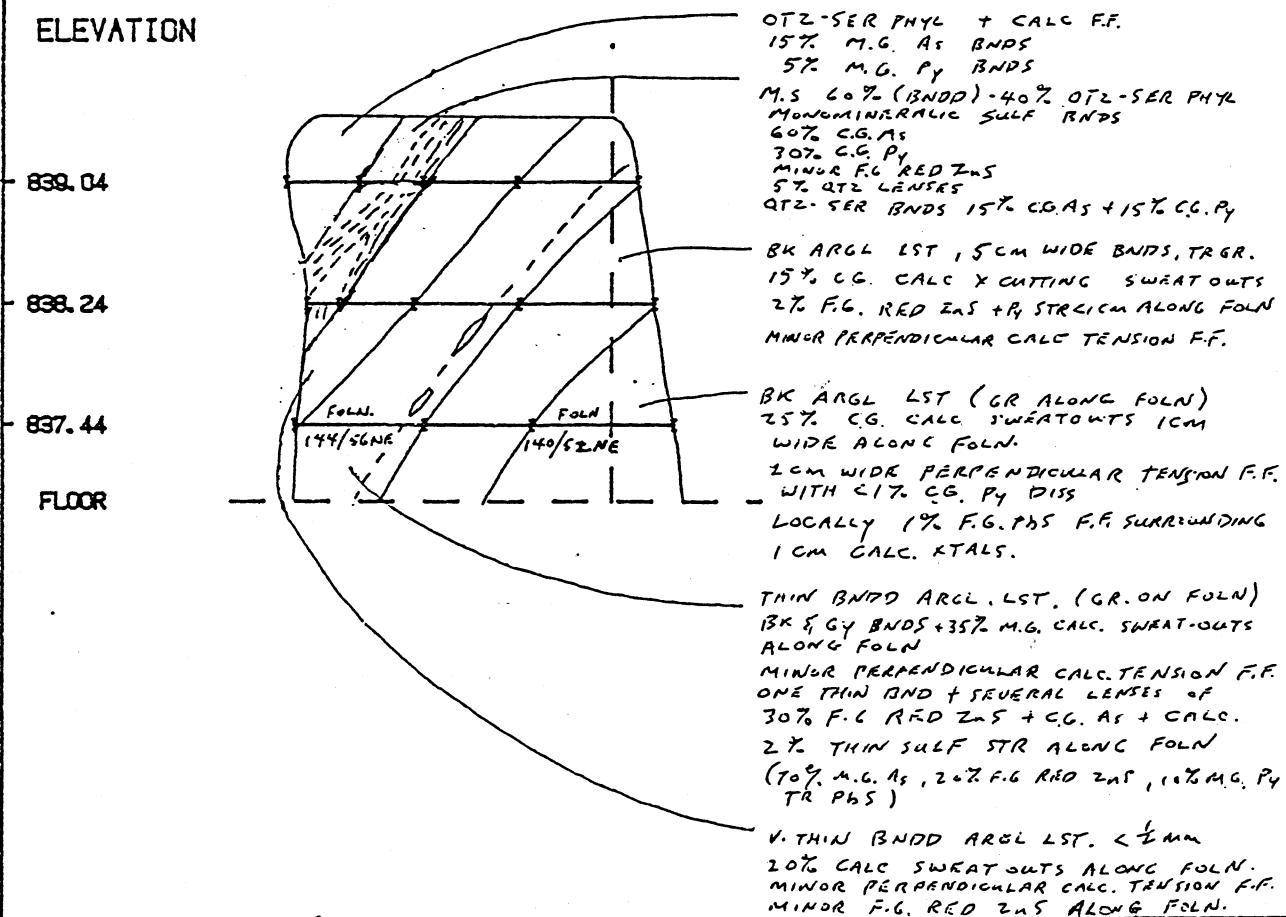
J & L PROJECT, B.C.

FACE# 111

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43135	.48	.10	.01	3.600	2.5	.9
43136	.43	2.00	.72	12.940	111.4	8.6
43137	.62	.41	.74	.427	6.8	.7
43138	.80	.06	.15	.648	.3	.5
43139	.21	.81	.74	3.470	39.0	3.8
43140	.50	.19	1.11	.111	20.2	.4
43141	.70	.09	.72	1.300	5.6	2.7
43142	.89	.06	.02	.017	3.8	.3

SAMPLE NUMBERS				GEOLOGY CODES			
049195	049196	049197	049198	07	12	02	02
049199	049190	049191	049192	12	02	02	02
049193	049194	049195		02	02	02	02

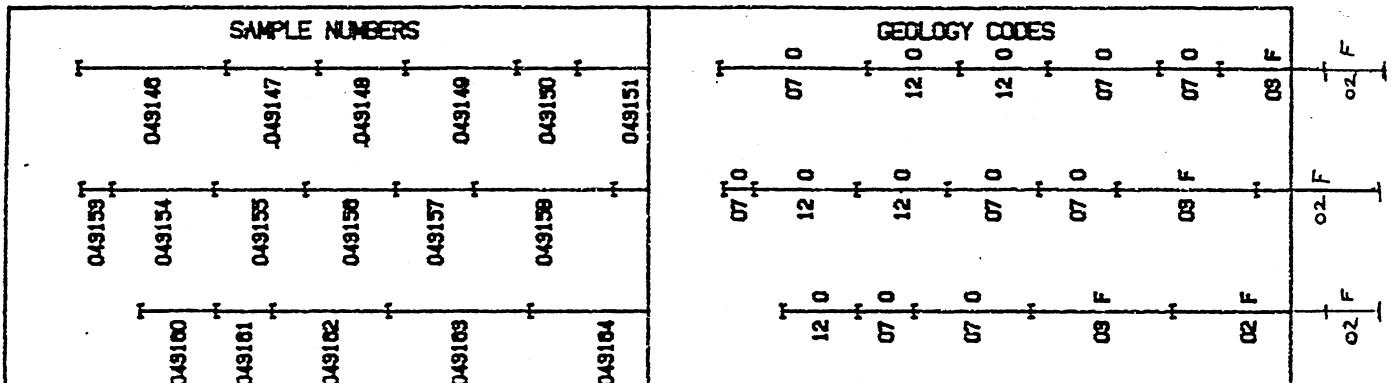
GEOLOGY



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

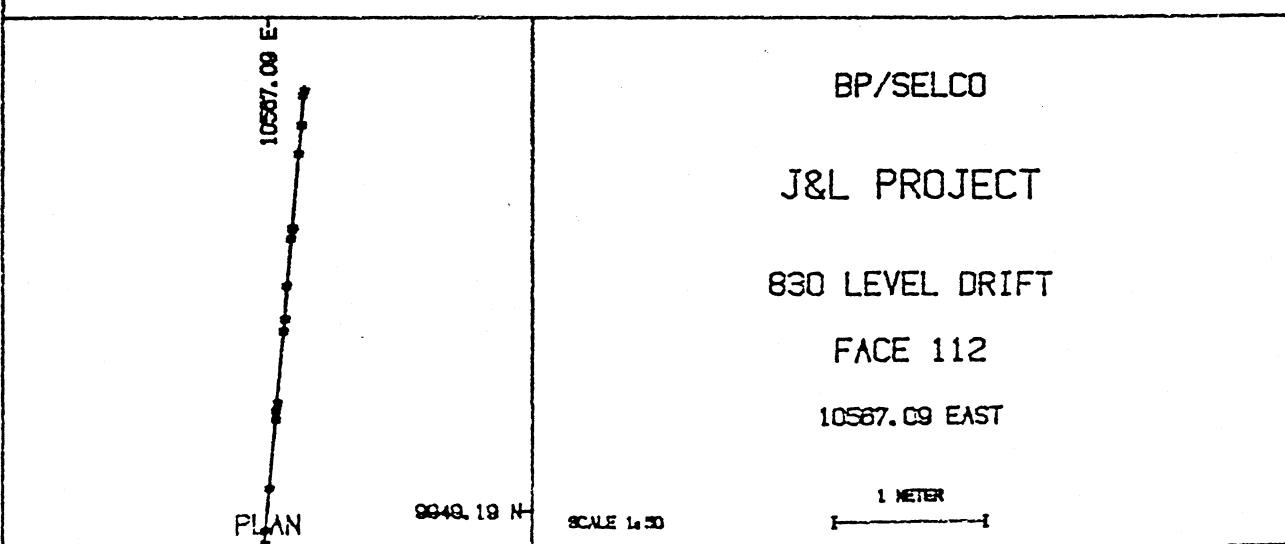
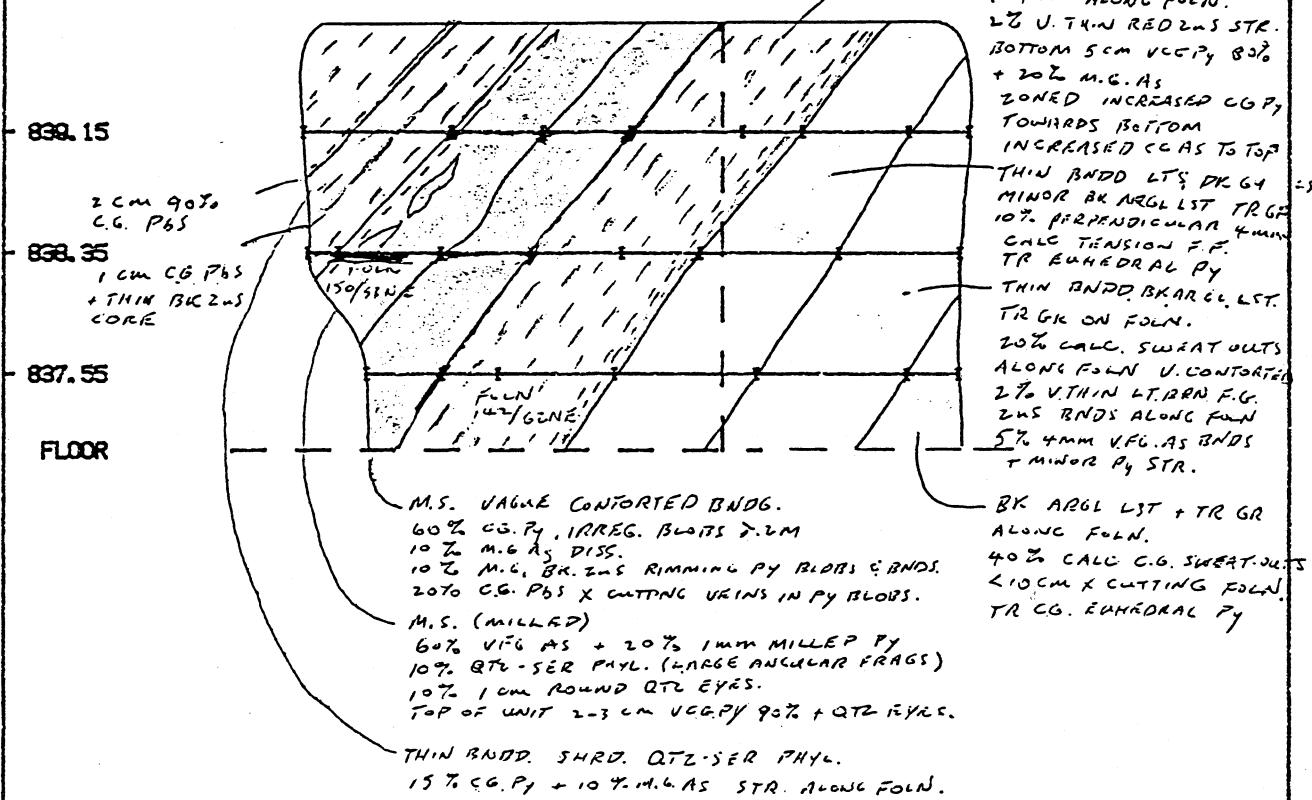
FACE# 112

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43146	.98	.39	.34	1.230	17.9	1.3
43147	.61	12.00	12.80	2.540	457.0	13.3
43148	.58	9.98	12.80	7.360	170.8	15.0
43149	.74	.31	.23	2.850	9.6	.7
43150	.40	.08	.05	.852	5.1	.4
43151	.70	.06	.15	.030	.3	.1
43152	.40	.22	.40	.392	7.3	.9
43153	.20	1.28	.04	5.780	16.1	5.8
43154	.68	12.80	17.60	2.300	515.6	13.0
43155	.60	10.30	9.39	6.630	321.7	16.3
43156	.60	.16	.07	1.330	7.0	.5
43157	.52	.28	.20	2.010	13.3	1.1
43158	.72	.14	.20	.176	8.4	.5
43159	.80	.32	.66	.828	16.2	1.6



GEOLOGY

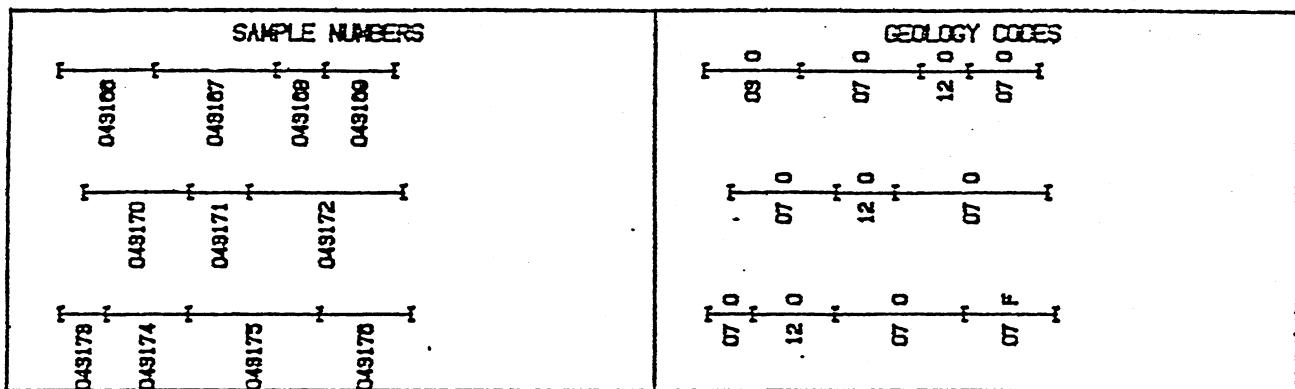
ELEVATION



SEILCO DIV. - DP RESOURCES J & L PROJECT, B.C.

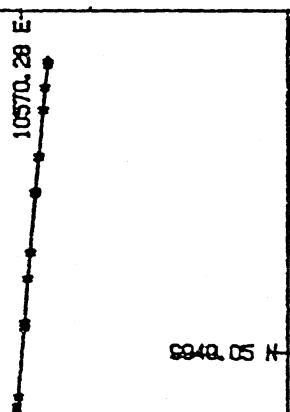
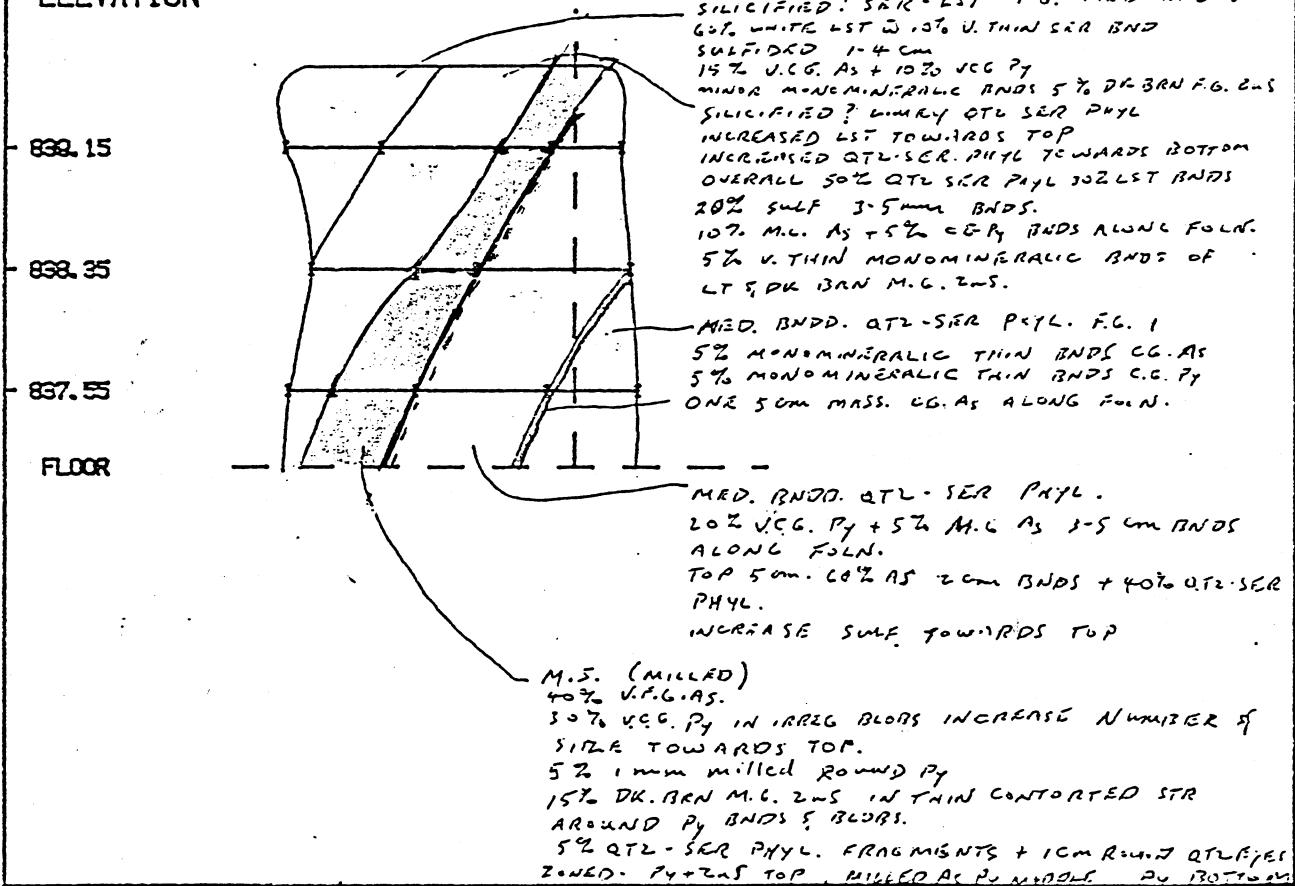
FACE# 113

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43166	.63	.06	.04	7.150	16.4	2.6
43167	.80	.13	.34	8.410	18.5	1.1
43168	.32	6.35	19.20	5.270	481.6	20.9
43169	.46	.33	.27	3.490	10.3	.8
43170	.70	.10	.10	4.330	6.8	1.6
43171	.39	9.58	16.30	5.270	225.9	10.7
43172	1.01	.33	.24	4.090	10.3	.8



GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

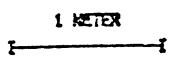
830 LEVEL DRIFT

FACE 113

10570.28 EAST

PLAN

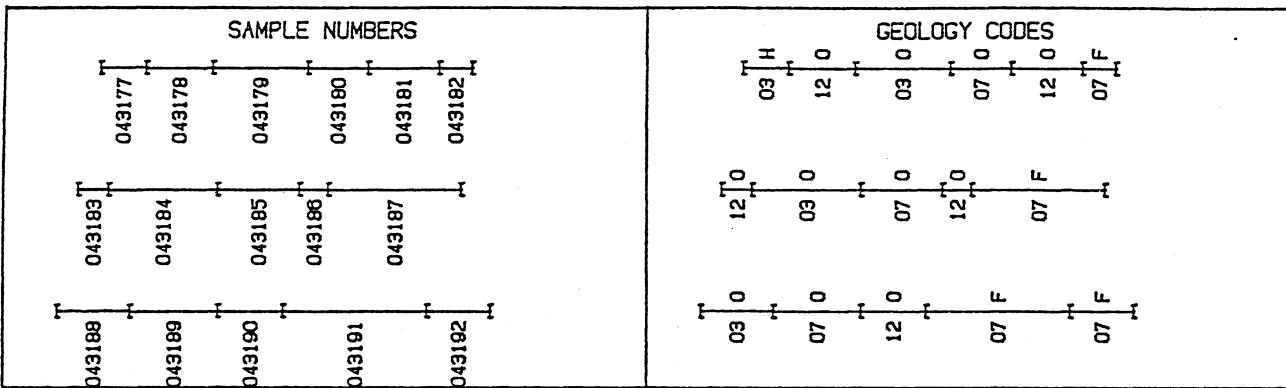
SCALE 1:50



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

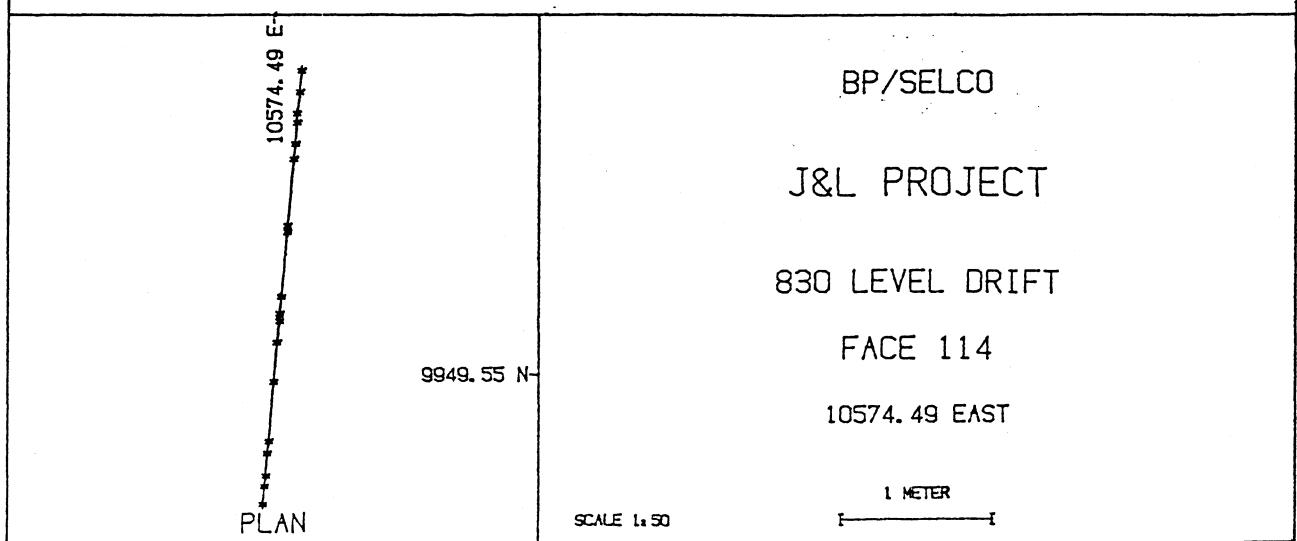
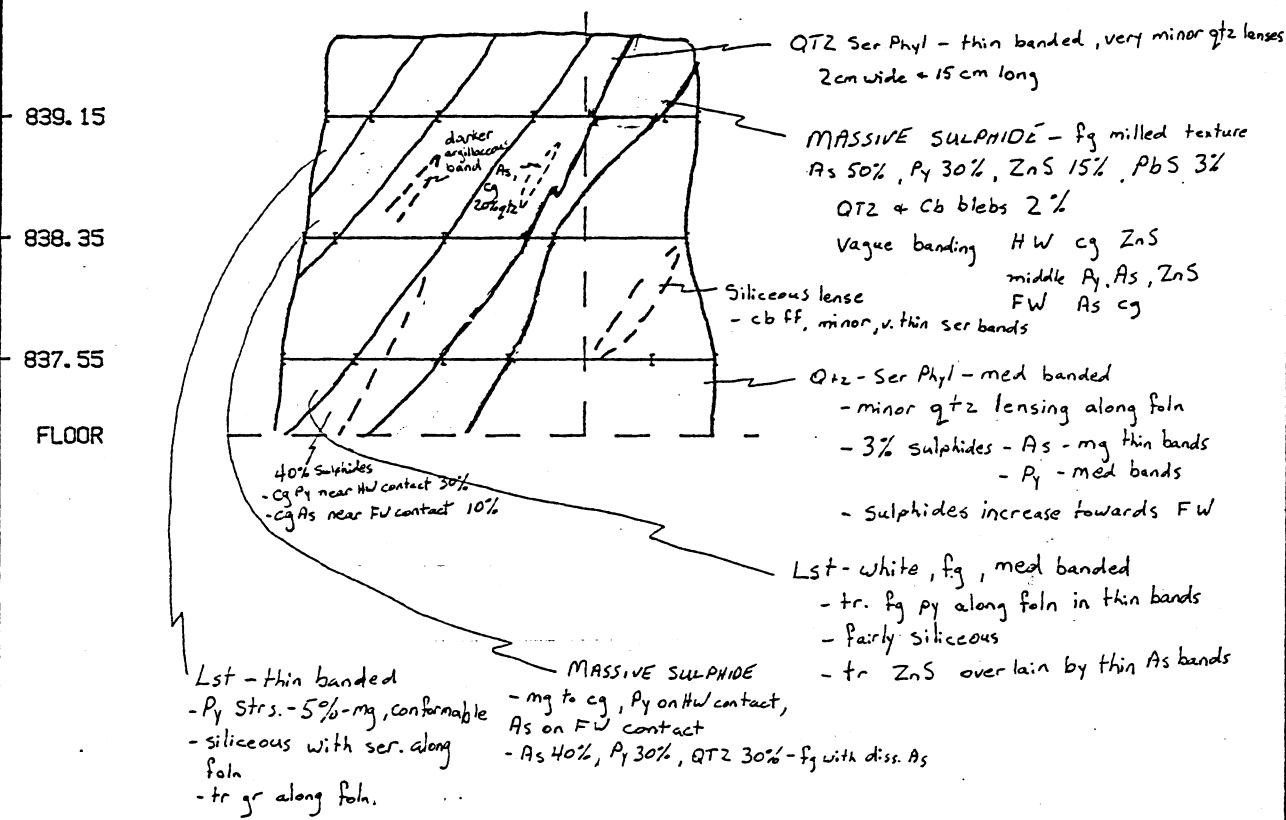
FACE# 114

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (o/mT)	GOLD (o/mT)
43177	.30	.05	.03	2.470	42.8	.6
43178	.44	.43	.04	14.400	110.0	1.7
43179	.63	.04	.02	2.940	10.6	.3
43180	.40	.14	.03	6.450	7.2	1.0
43181	.47	7.41	18.60	8.240	206.0	12.8
43182	.22	.11	.17	2.820	5.5	.1
43183	.20	.10	.03	18.900	22.3	1.9
43184	.72	.06	.03	.877	6.8	.2
43185	.54	.48	1.04	9.660	8.9	1.3
43186	.19	6.85	14.40	10.500	144.3	9.2
43187	.88	.23	.25	5.410	7.2	.3



GEOLOGY

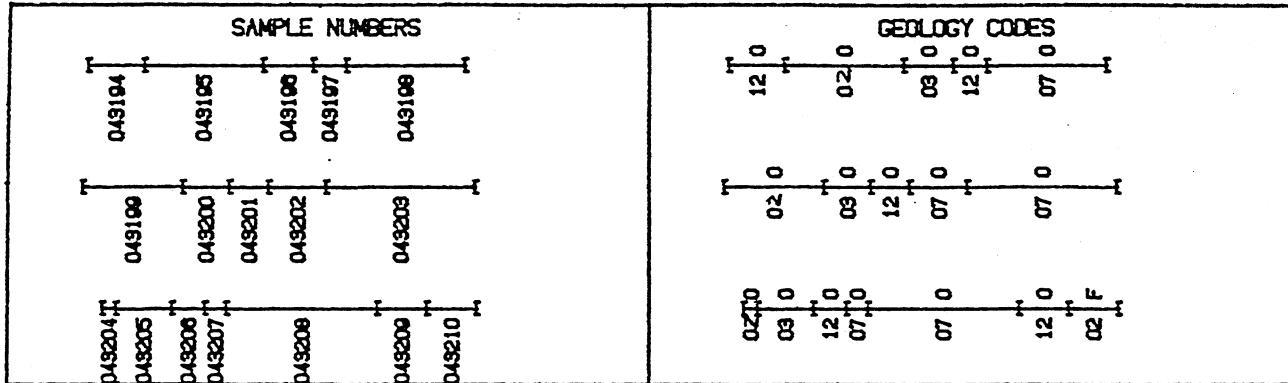
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

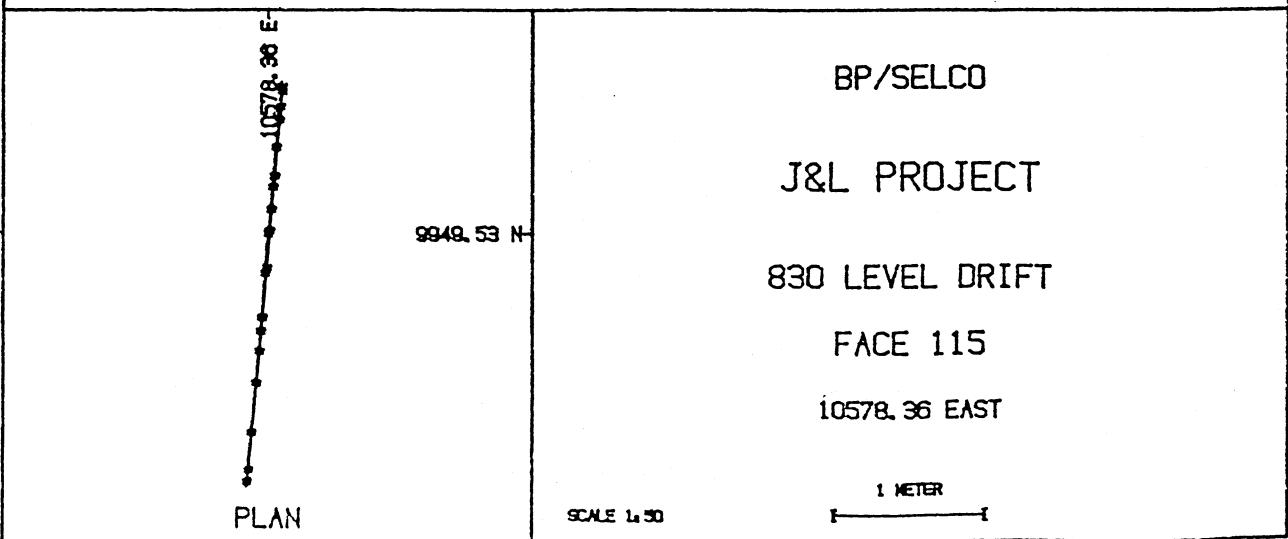
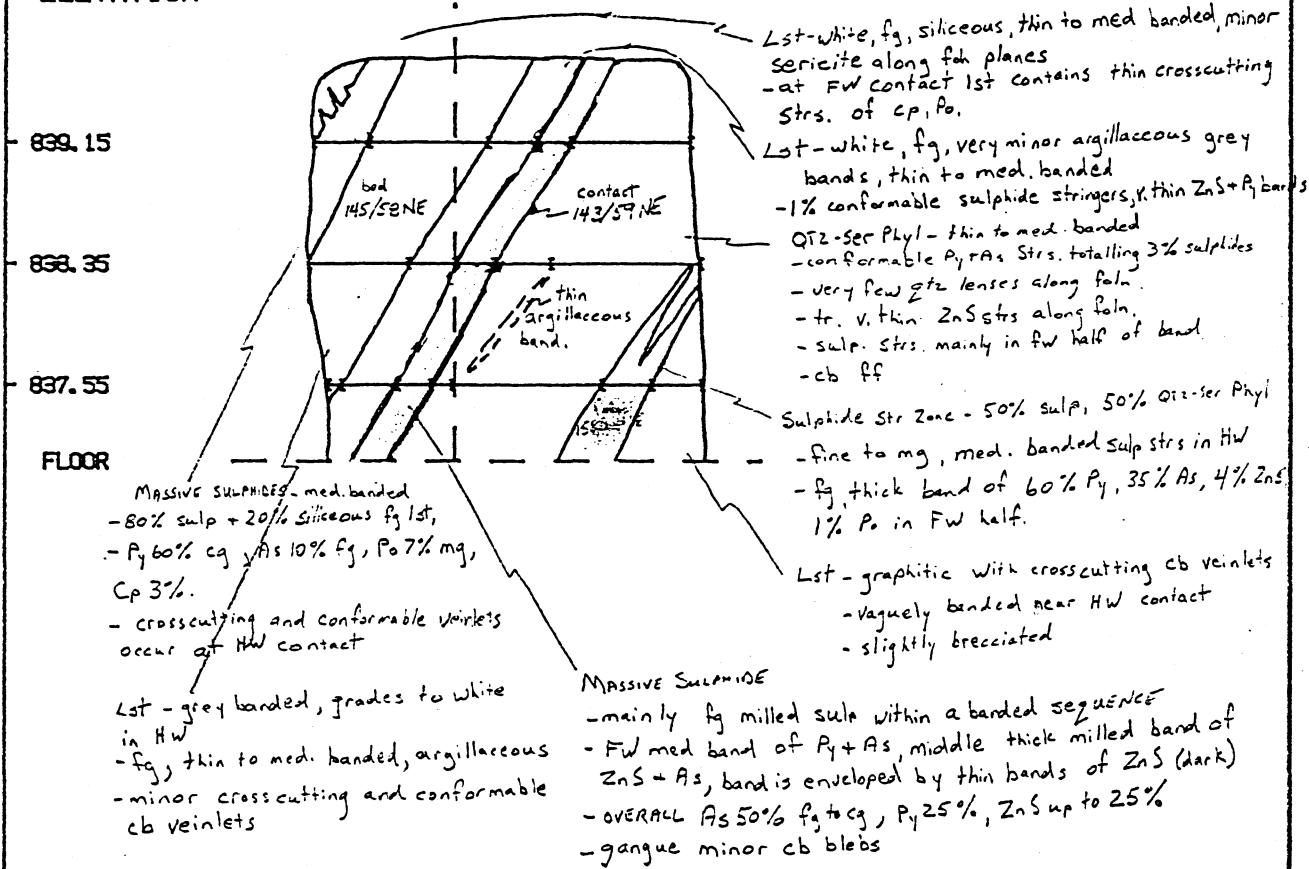
FACE# 115

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43194	.37	.27	.04	3.330	2.7	.5
43195	.79	.01	.19	.139	3.1	.2
43196	.33	.50	.84	1.370	7.8	.9
43197	.22	6.86	14.40	8.270	154.6	13.3
43198	.79	.08	.06	1.770	5.1	.3
43199	.66	.02	.09	.045	3.1	.1
43200	.31	.08	1.22	1.450	3.4	.9
43201	.26	7.17	13.90	5.870	341.7	17.4
43202	.38	.41	.45	5.230	13.7	1.3
43203	.99	.08	.03	1.430	11.6	.1



GEOLOGY

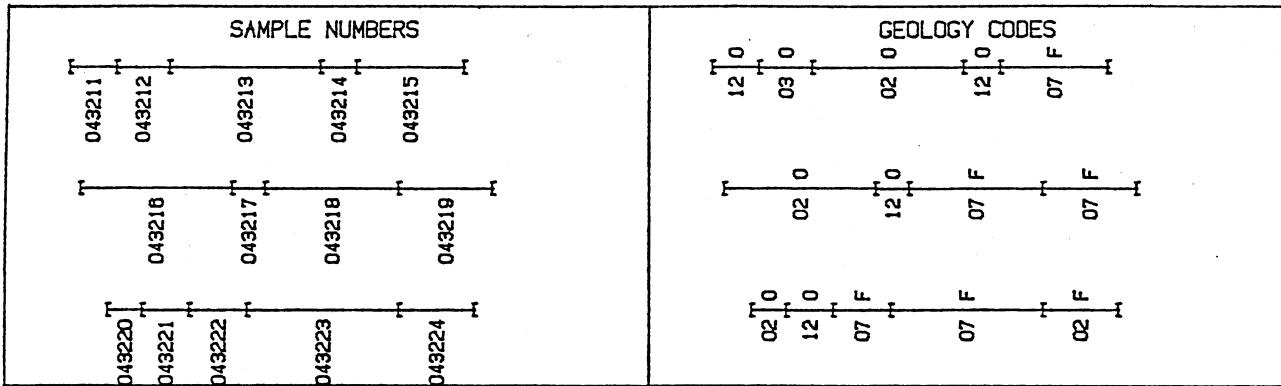
ELEVATION



SELCO DIV. -- BP RESOURCES J & L PROJECT, B.C.

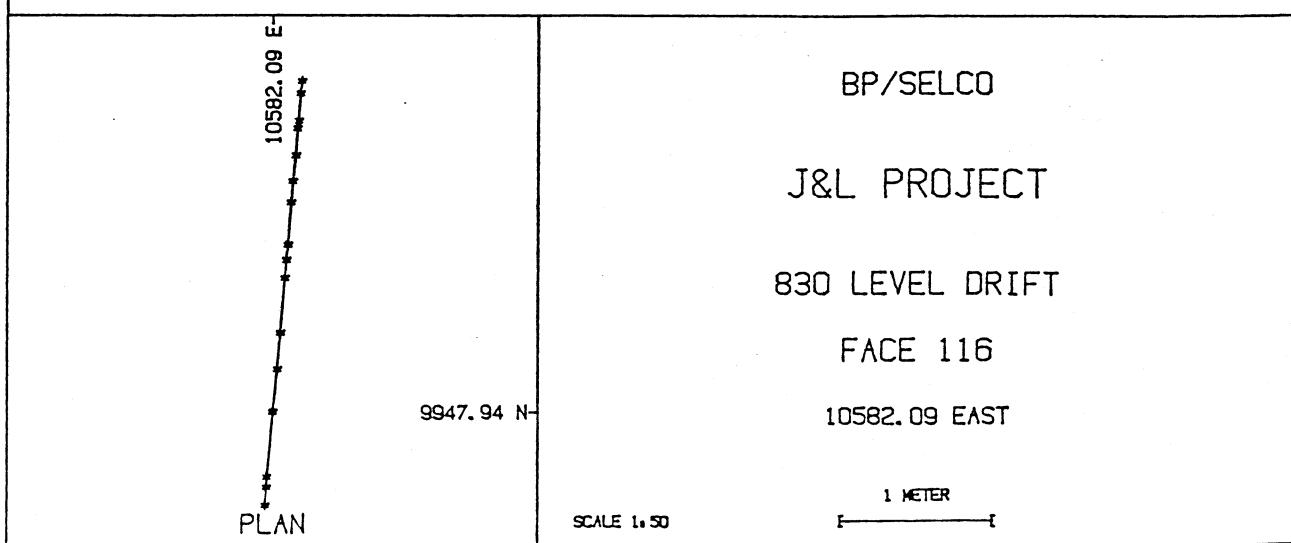
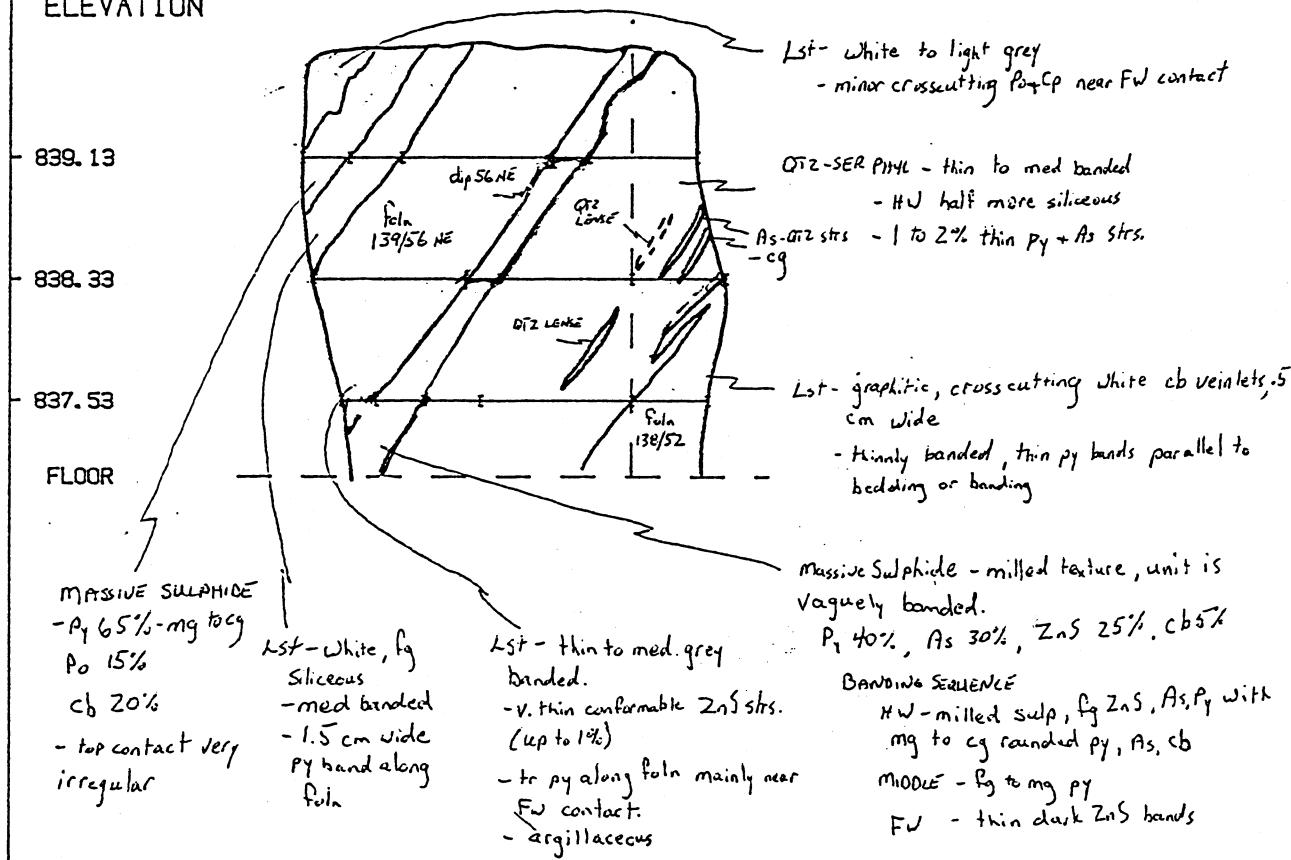
FACE# 116

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43211	.31	.01	.01	4.670	.4	1.0
43212	.35	.02	.01	.079	.3	.4
43213	1.00	.08	.12	.353	.3	.5
43214	.24	6.65	9.03	4.860	129.9	14.1
43215	.71	.15	.04	4.900	.3	1.5
43216	1.00	.37	.94	.887	43.5	2.4
43217	.22	7.92	9.58	17.770	160.1	25.0
43218	.88	.13	.08	1.190	.3	.3
43219	.62	.66	.08	8.500	17.2	2.0
43220	.23	.31	.90	2.400	7.4	2.9



GEOLOGY

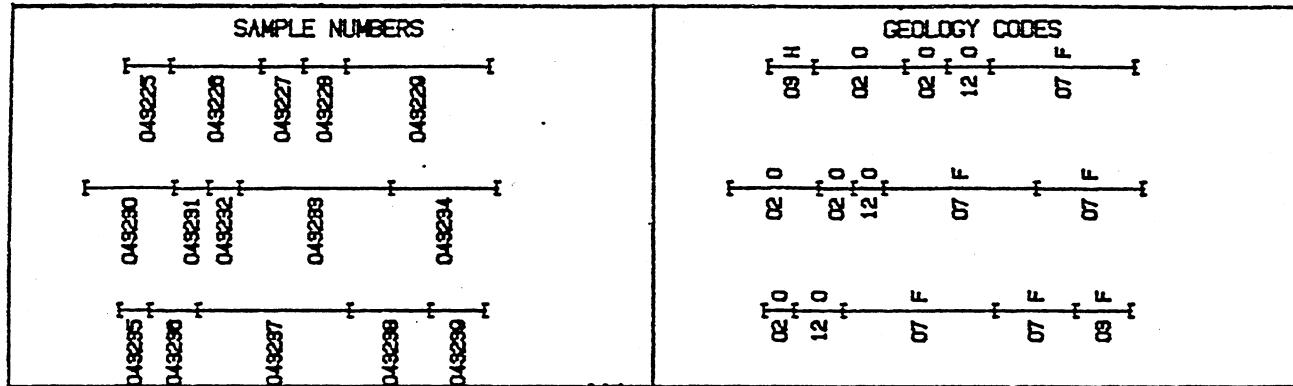
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 117

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43225	.30	.05	.46	.023	.3	.5
43226	.60	.62	7.93	4.950	7.1	.5
43227	.28	1.14	1.64	1.500	6.8	2.1
43228	.28	7.89	16.60	.076	136.6	14.2
43229	.95	.16	.22	4.160	3.1	.3
43230	.59	.77	13.90	.046	10.0	.3
43231	.23	.56	.79	.367	7.9	1.0
43232	.20	7.61	16.30	5.180	160.2	16.0
43233	1.00	.09	.06	2.860	4.1	.7
43234	.70	.03	.02	1.890	.3	.8



GEOLOGY

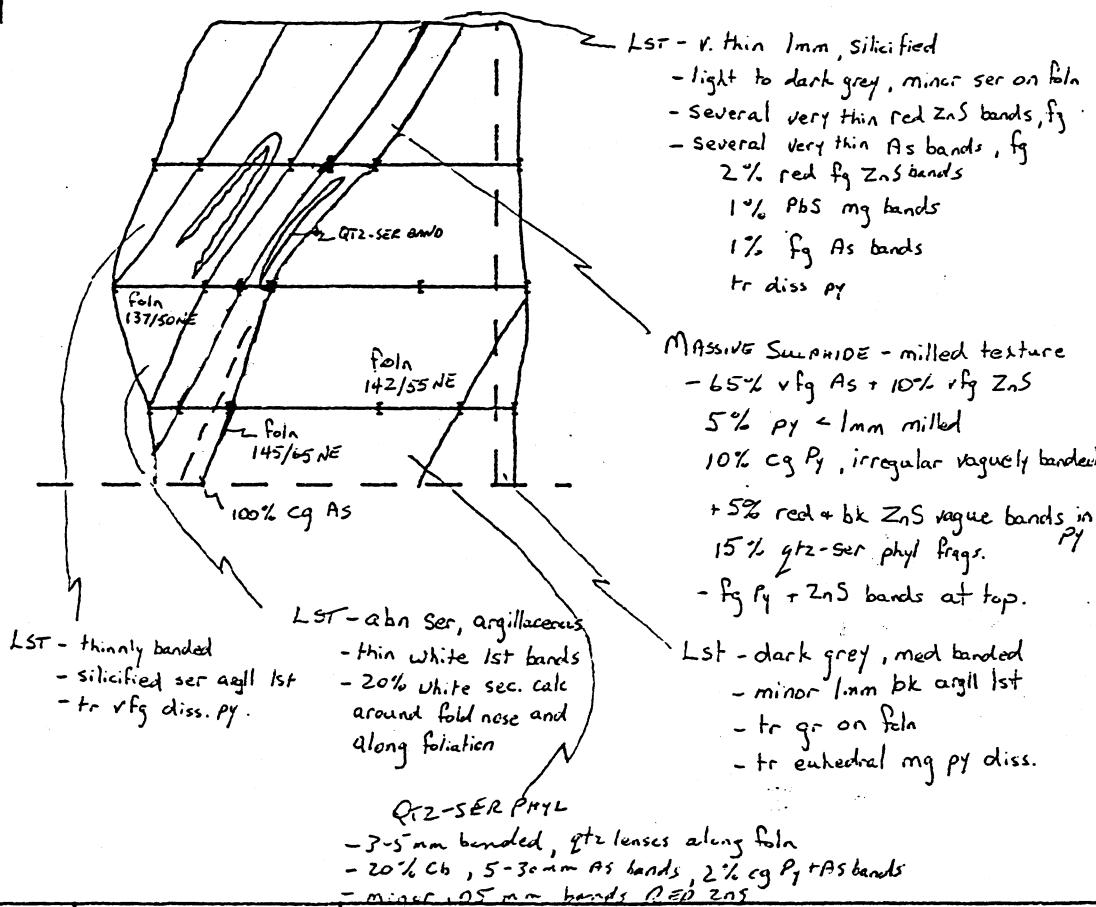
ELEVATION

839.13

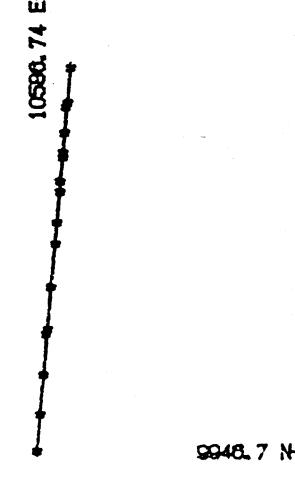
838.33

837.53

FLOOR



10586.74 E



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 117

10586.74 EAST

1 METER

PLAN

SCALE 1:50

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 118

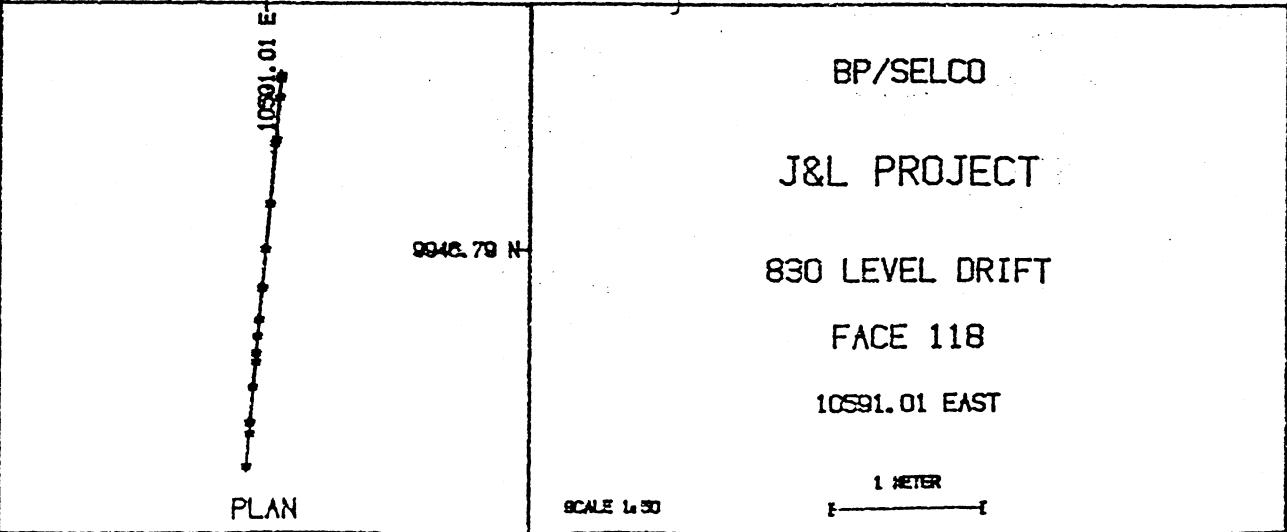
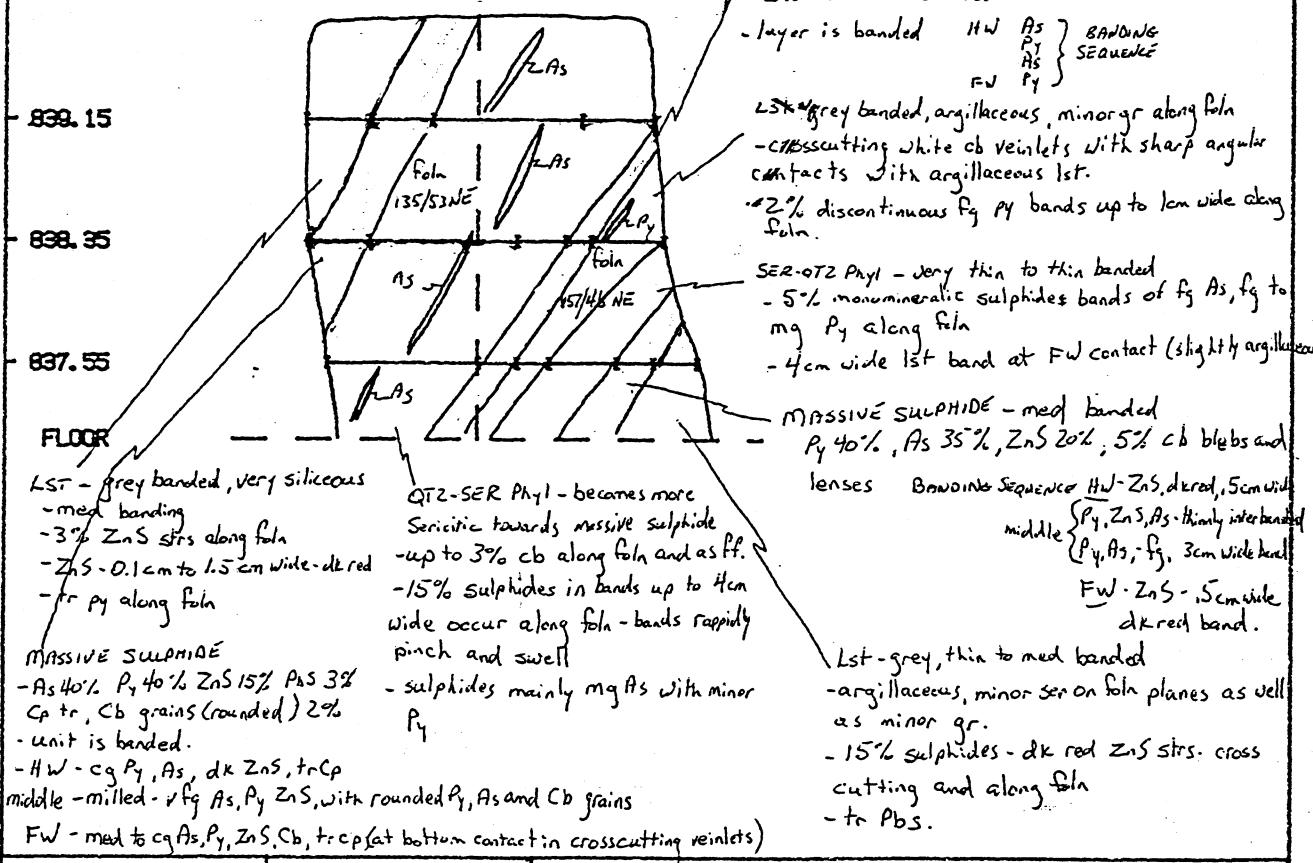
*** *** *** *** ***

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43241	.44	.13	2.74	.285	11.0	.6
43242	.40	7.54	8.86	13.780	169.3	30.9
43243	.99	.12	.09	5.310	1.8	3.7
43244	.47	.06	.02	10.300	5.5	13.0
43245	.40	7.36	11.50	6.690	142.9	20.3
43246	.98	.15	.17	7.800	8.3	10.9
43247	.32	.03	.03	5.940	3.3	.6
43248	.17	.33	.02	14.490	39.7	3.5
43249	.48	.10	.02	.486	1.7	1.0

SAMPLE NUMBERS				GEOLOGY CODES					
049211	049212	049213	049214	H 0 0 R 5 5					
049215	049216	049217	049218	0 0 0 0 0 0					
049220	049221	049222	049223	0 0 0 0 0 0					

GEOLOGY

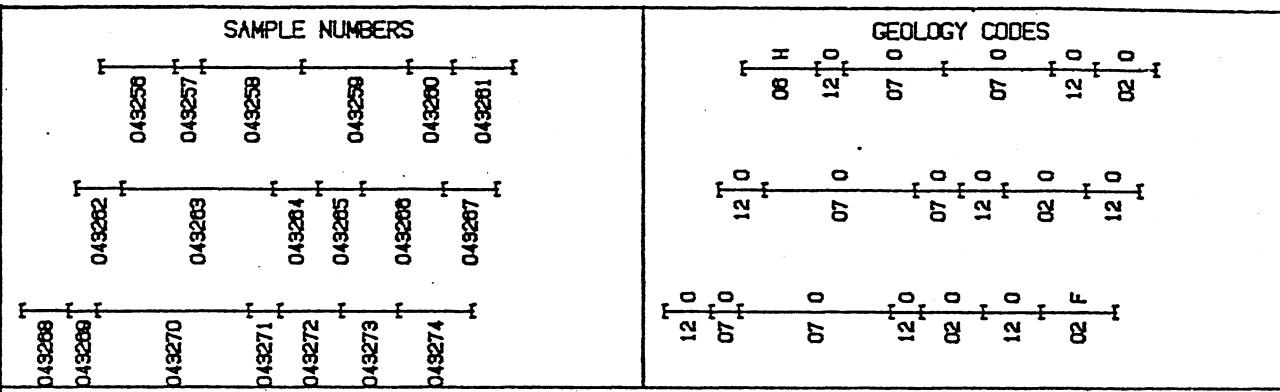
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

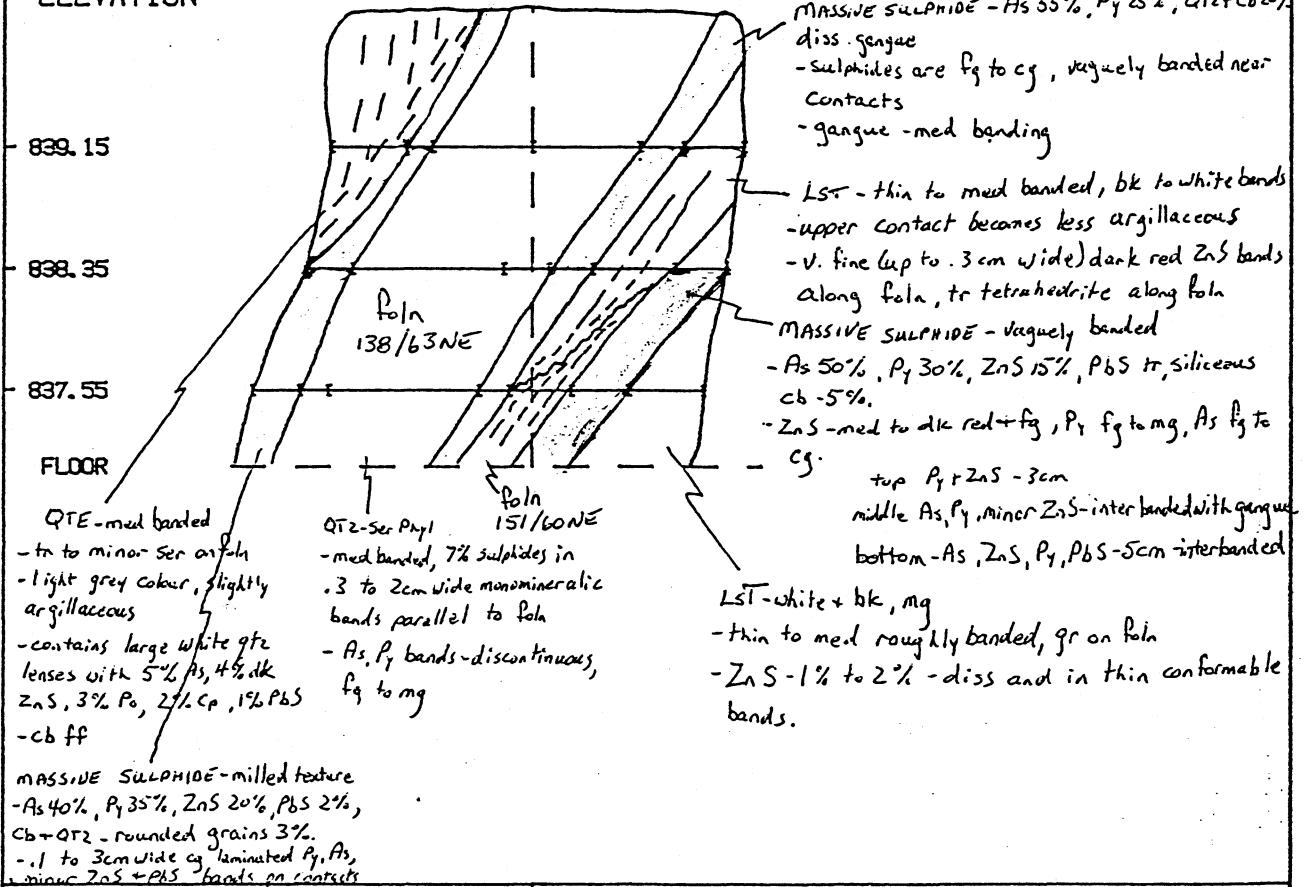
FACE# 119

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43256	.49	.03	.25	.192	1.6	.4
43257	.18	6.25	11.30	11.750	105.1	14.9
43258	.66	.15	.10	1.515	1.9	.8
43259	.71	.04	.02	5.350	.3	1.8
43260	.29	.25	.01	15.950	23.0	3.1
43261	.40	.19	.24	5.780	5.3	5.7
43262	.30	9.08	16.80	5.410	159.0	14.4
43263	1.00	.24	.26	3.230	2.5	1.6
43264	.30	.03	.02	6.330	.3	1.7
43265	.29	.15	.01	20.560	7.8	2.5
43266	.51	.12	.13	2.850	1.2	1.5
43267	.35	4.48	5.31	13.020	94.6	15.6



GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 119

10594.56 EAST

10594.56 E
9948.1 N
FLAN

SCALE 1:50

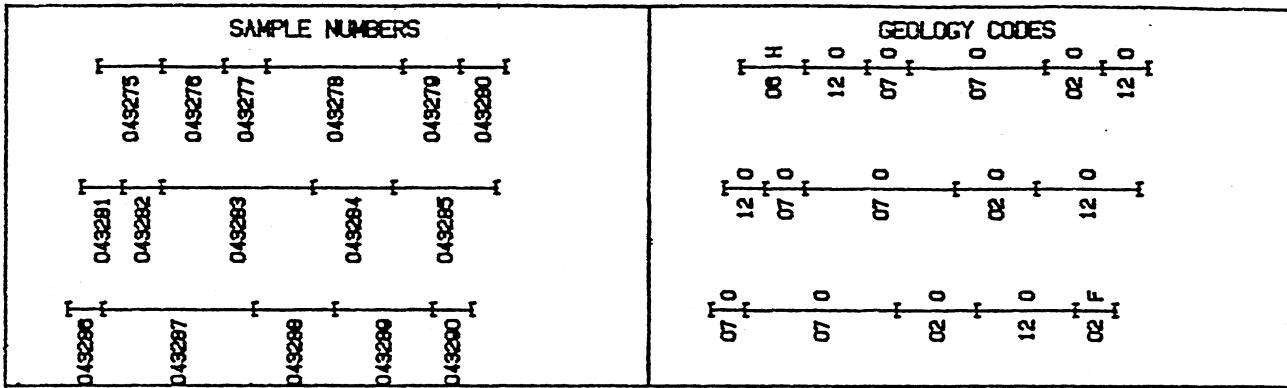
1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 120

* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43275	.42	1.53	.77	.099	23.8	.9
43276	.41	5.16	8.86	14.300	121.5	38.9
43277	.28	.14	.15	.355	7.4	.8
43278	.90	.26	.04	5.940	15.8	1.3
43279	.39	.12	2.70	1.220	8.3	1.2
43280	.30	1.42	3.53	7.520	32.6	7.2
43281	.27	7.09	11.70	5.210	111.1	13.7
43282	.26	.34	.38	3.850	9.6	3.4
43283	1.00	.32	.13	8.790	24.2	1.8
43284	.53	.18	1.15	.481	41.0	.8
43285	.60	3.11	4.43	13.400	96.8	23.9



GEOLOGY

ELEVATION

- 838.15

milled
vaguely
banded

contact
141/63 NE

MASSIVE SULPHIDE

- horizon is banded
Hw - cg, vaguely banded and contorted, 3 to 10 cm wide
As 70%, Py 10%, ZnS 6% (dk red), PbS 1%, C 1%
Qtz 12%

- 838.35

contact
148/66 NE

milled - milled texture, 15 to 20 cm wide
As 40%, Py 40%, ZnS 20%

- v fg with mg to cg rounded As + Py
Fw - same as Hw

Banding Sequence
As
ZnS
milled texture
ZnS
As

- 837.55

FLOOR

QTE - clean, med banding

- tr ser on foln planes
- 4% sulp str. ~ 1 cm wide
- not monomineralic bands
- ZnS - dk red, Py + PbS 10%

QTZ-SEL PYRL

- thin to med banded, minor gtz lenses
- more ser. towards Hw
- higher sulp. occur towards Fw
- 10% - 15% sulp. - mainly monomineralic
- mg As bands up to 2 cm wide
- minor fg to mg Py bands up to 2 cm wide

LST

- white to grey, thin to med - argillaceous
- very minor (4 cm wide) gtz-ser band at Fw
with 4% Py, 3% As

- 2 to 5 cm wide dk red ZnS band directly above gtz-ser band, ZnS - fg
- tr diss PbS throughout

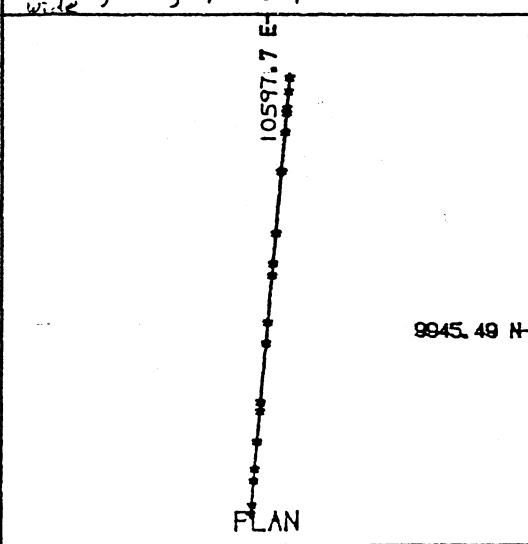
MASSIVE SULPHIDE

- Py 40%, As 35%, ZnS 20%, PbS 3%, Cb 4%
- top 15 cm thin to med banded - ie monomineralic
Py + As bands (50:50) with 20% gtz + minor ser.

- massive band is fg to mg - irregular and
contorted, no banding + texture
- cb occurs as cg diss. blebs.

Lst - cg, roughly banded, argillaceous
- gr on foln planes

- v. minor diss. dk red ZnS
- tr tetratedrite with minor
PbS
- total sulp. ≈ 2%



BP/SELCO

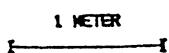
J&L PROJECT

830 LEVEL DRIFT

FACE 120

10597.70 EAST

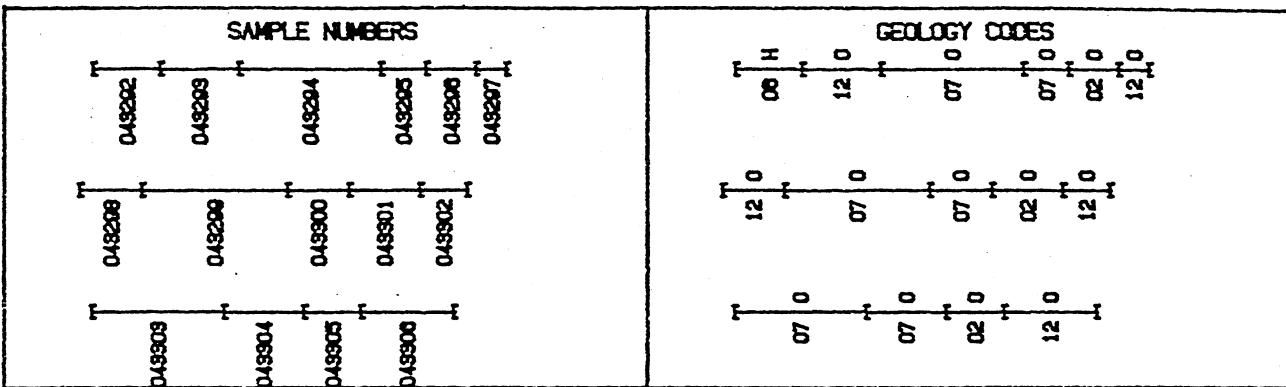
SCALE 1:50



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 121

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (o/mT)	GOLD (o/mT)
43292	.44	.04	.04	5.200	.3	.3
43293	.52	1.60	3.05	21.100	51.4	49.4
43294	.94	1.08	.43	.136	29.7	6.6
43295	.30	.19	.02	5.790	80.1	3.5
43296	.33	.10	.23	1.340	5.7	1.1
43297	.20	1.34	2.70	15.600	67.9	21.2
43298	.41	3.23	12.90	14.700	69.6	33.2
43299	.96	1.43	1.30	6.660	37.5	7.7
43300	.41	.57	.06	8.450	49.3	3.5
43301	.47	.06	.24	.975	2.1	1.3
43302	.31	.52	.27	15.700	54.3	14.3



GEOLOGY

ELEVATION

- 839.28

- 838.48

- 837.68

FLOOR

QTE - med bndd

- tr Ser on foln

- tr Sulp - fg Py on foln

- tr diss dk red ZnS

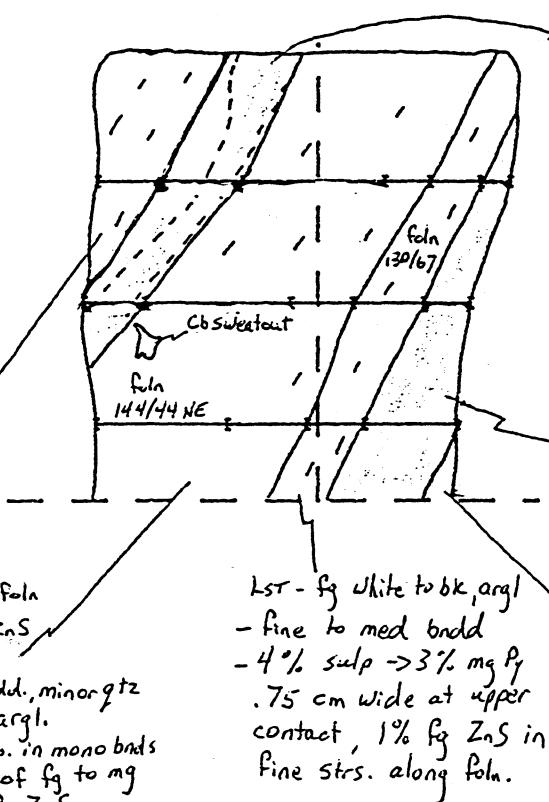
QTZ-SER PHYL

- fine to med bndd, minor qtz
lenses, slightly argl.

- up to 10% sulp. in mono bndds
of 0.1 to 3.0 cm of fg to mg
Py, fg to mg As, fg ZnS.

- Py 4%, As 4%, ZnS 2%

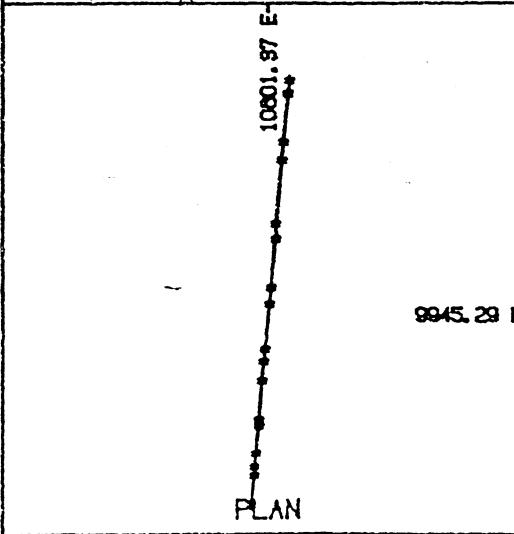
- sulp. more coars. at lower
30 cm and upper 10 cm.



M.S. - As 50%, Py 25%, ZnS 15%, Qtz 10%
- unit is banded
- top 5 to 20 cm roughly bndd
 cg Py 15%
 cg As 80%
 dk red ZnS 5%
 + tetrahedrite in Cb clots
- middle - v.fg. milled texture, 4 to 10 cm
 fg ZnS + Py + As, rounded mg Py + As
- bottom - cg. As 75%, Py + ZnS 15%, Qtz
 + Cb clots 10% - not banded

M.S. - As 50%, Py 25%, ZnS 20%, Qtz 5%
- massive, fg to mg
- upper 10 cm very vaguely bndd - fine
 to med bndd with As, Py, Qtz
- higher ZnS coarsen at lower 10 cm.

LST - roughly bndd
- blk + white, argl
- gr - minor along banding.



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 121

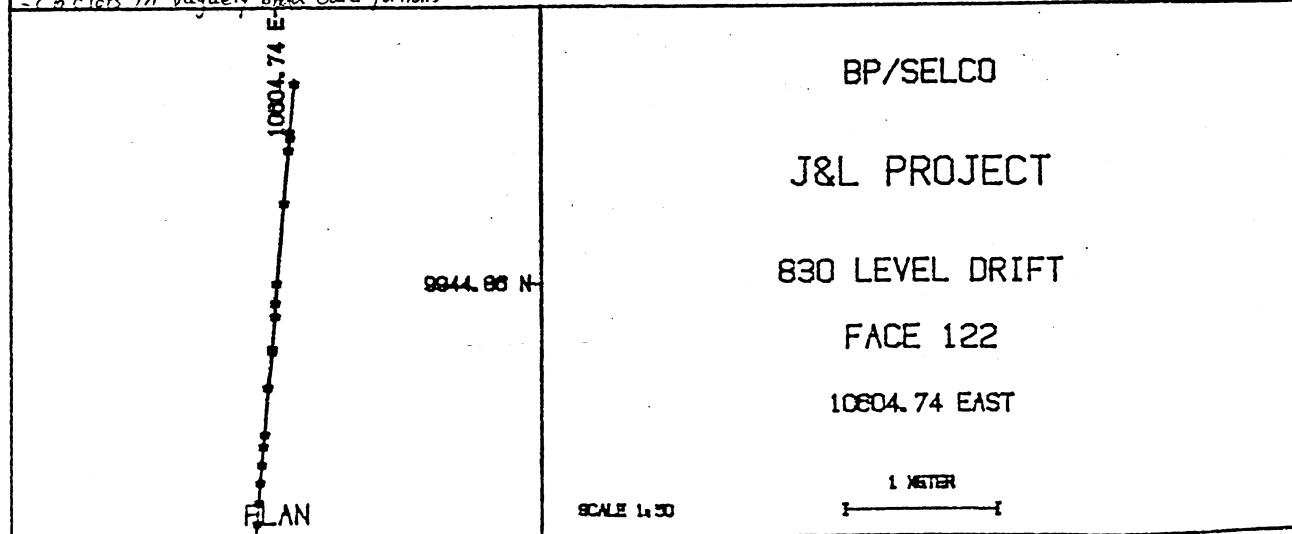
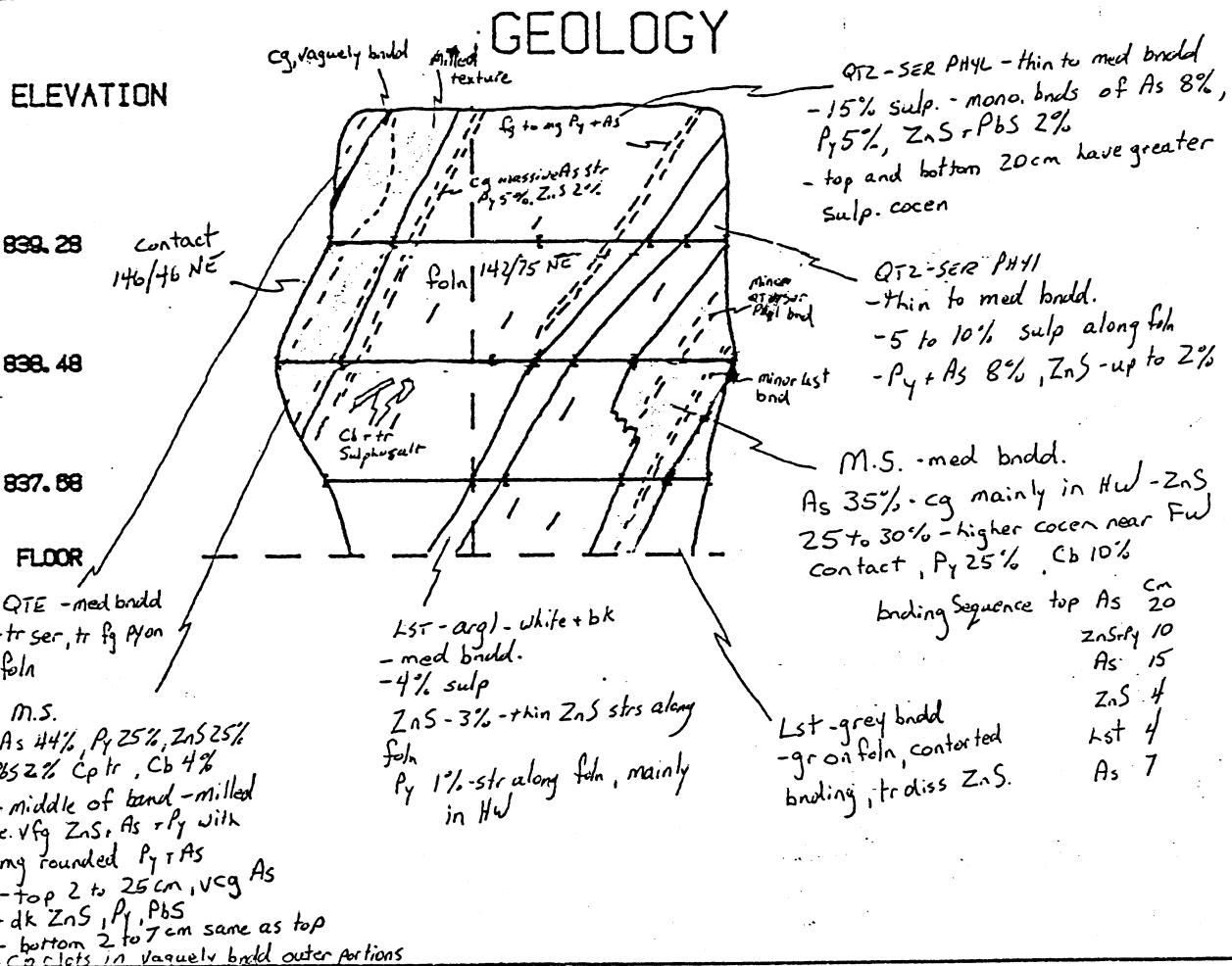
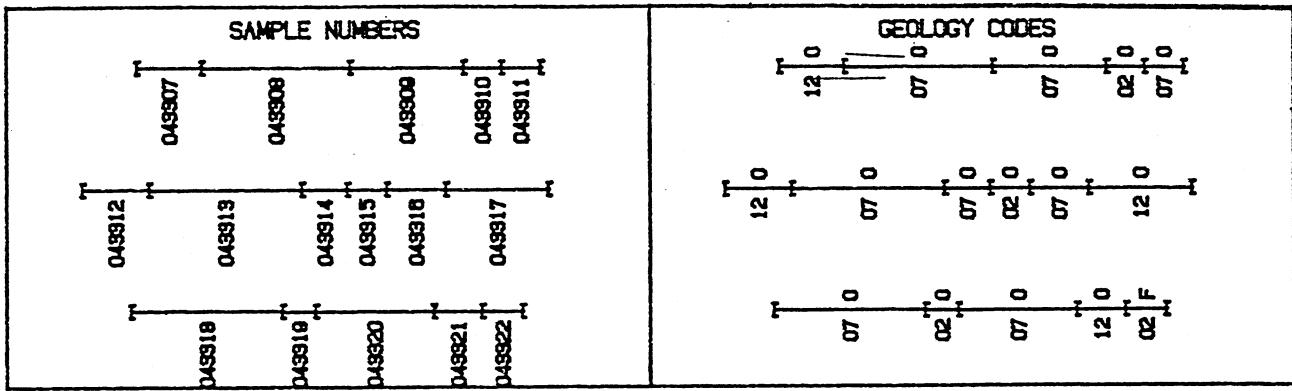
10601.37 EAST

1 METER

SELCO DIV. - DP RESOURCES J & L PROJECT, B.C.

FACE# 122

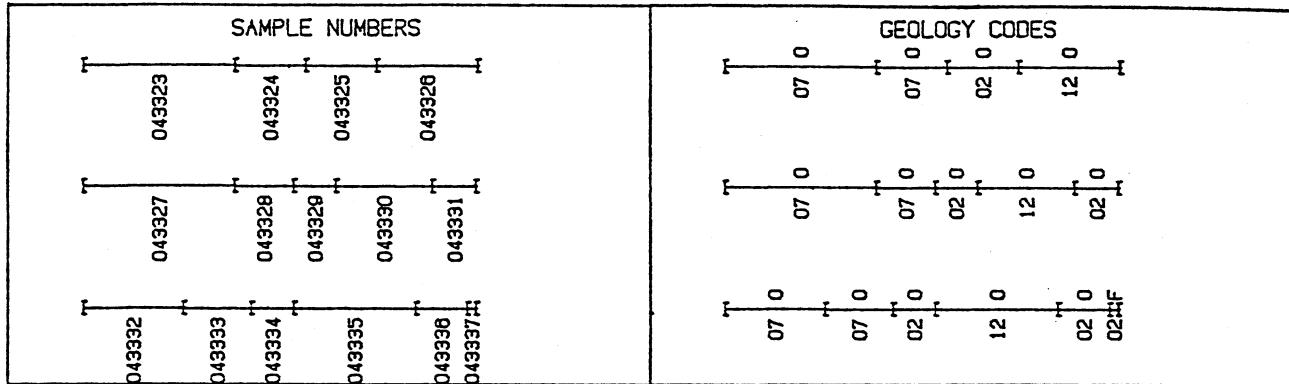
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43307	.43	2.56	8.52	17.200	101.2	40.0
43308	.98	.50	.14	4.060	14.4	4.8
43309	.75	.63	.14	6.380	72.0	4.1
43310	.25	.16	.12	.960	10.6	.4
43311	.26	.42	.02	1.740	28.1	1.4
43312	.44	3.47	9.60	14.900	80.3	31.4
43313	1.01	.76	.92	9.240	2.9	12.9
43314	.30	1.12	.04	8.940	116.9	5.1
43315	.26	.06	.16	.193	.3	.6
43316	.39	.24	.34	2.710	13.9	1.9
43317	.68	1.28	4.52	12.400	44.2	12.7



SELCO DIV. - BP RESOURCES J & L PROJECT. B.C.

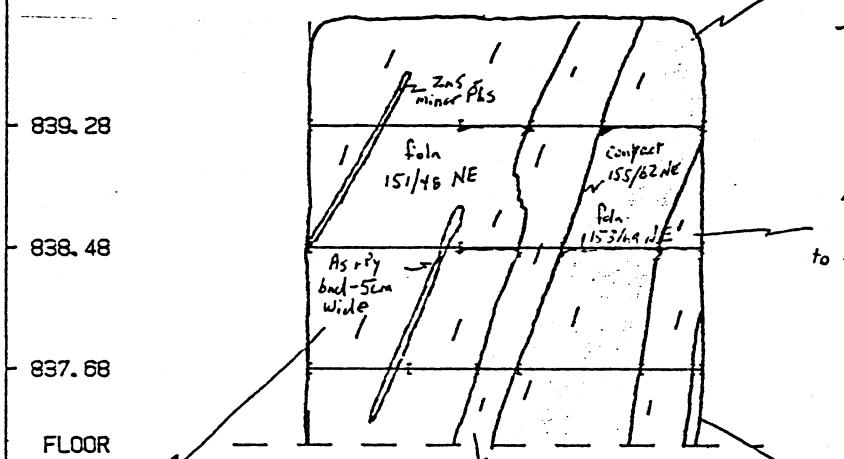
FACE# 123

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43323	1.00	.41	.74	2.300	25.4	.8
43324	.47	.04	.01	9.780	4.8	6.2
43325	.47	.10	.21	2.060	.3	1.0
43326	.67	.40	.10	10.200	.7	6.0
43327	1.00	.10	.01	3.680	5.7	2.5
43328	.39	1.17	1.54	13.000	23.0	13.5
43329	.28	.07	.14	.717	.3	.6
43330	.64	.79	.35	13.800	4.8	6.8
43331	.29	.97	2.87	7.420	14.4	11.5



GEOLOGY

ELEVATION



QTZ-SEL Phyl
 - thin to med bndl gte lenses along
 - 15 to 20% sulp mainly within lower 60 cm as mono. massive bnds parallel to feln. - bands are from .2 cm to 5 cm wide.
 - band compositions are of
 - band compositions are of
 - fg As 6%, fg Py 5%.
 - one ZnS + PbS bnd occurs approximately 1.10 meters from bottom contact and is 4 cm wide

Lst - white to grey, med bndl, fg - very slightly argl. and siliceous - 5% Sulp. - Py 3%, ZnS 2%
 - Py 5trs occur parallel to feln main in the top 15cm of the unit
 - Py occurs in mono. fg bands, 1 to 2 cm wide
 - ZnS occurs in thin mono. bnds at lower 15cm parallel to feln.

M.S. - As 50%, Py 30%, ZnS 5%, tr tetrahedrite siliceous lst + ser 15%
 - top 45 cm med. bndl with fg to mg mono. bnds of Py 50%, As 30% + 20% gte + ser.

Lst - med. bndl, argl. r 45% sulp.
 - Py 25%, As 15%, ZnS 5%
 - Sulp. occur in mono. bnds parallel to feln - Py - 5cm wide bndl at FJ contact - As - 3cm wide bndl in middle of the unit - ZnS - in thin multiple bnds within the top 10 cm of the unit

Lst - thin to med bndl, white to grey, argl. to gr on feln
 - minor fg ZnS occur along feln in bnds up to .5 cm wide.

- the band is fg and very vaguely banded with dk red ZnS and red-brn ZnS with minor PbS.

9944.46 N

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 123

10607.92 EAST

PLAN

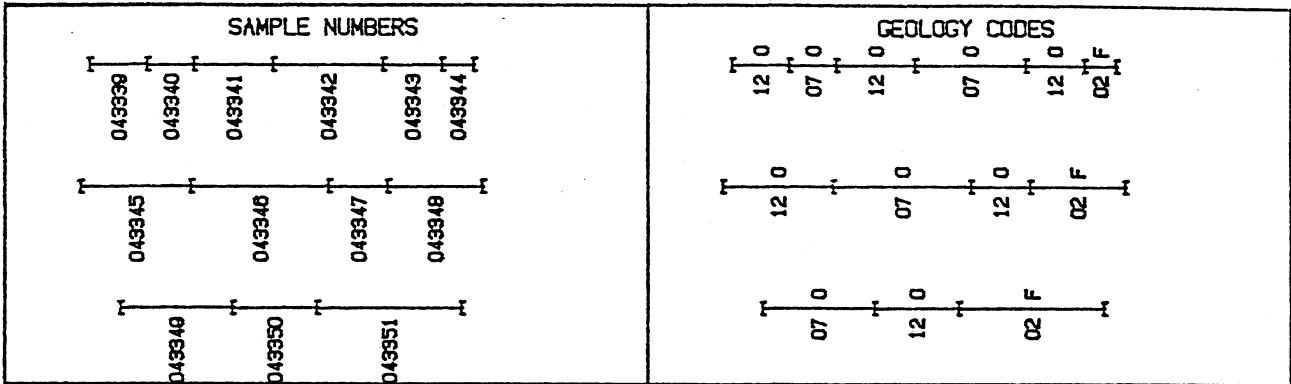
SCALE 1:50

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 124

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43339	.38	.15	.03	8.070	1.7	6.5
43340	.31	.10	.01	2.210	1.7	1.2
43341	.52	.21	.01	18.300	7.2	6.5
43342	.73	.15	.28	1.950	6.9	1.9
43343	.39	.52	.52	14.300	2.7	6.2
43344	.21	2.06	1.86	11.800	29.1	16.1
43345	.73	.25	.02	14.300	1.0	3.7
43346	.91	.20	.04	3.610	5.5	2.7
43347	.39	.72	.69	10.600	23.0	11.4
43348	.63	5.42	3.53	2.860	137.1	4.2



GEOLOGY

ELEVATION

- 839.32

- 838.52

- 837.72

FLOOR

QT2-SER Phyl
- thin to med bndl
- Sulf 20%
As 15% Py 10% in
med mono. bndl parallel
to foln.
- tr ZnS

M.S. - As 45%, Py 15%, Cpt.,
2t2 + minor ser 40%
- unit is med bndl parallel
to foln. in mono. bndl.
- Py occurs mainly within bottom
10 cm

QT2-SER Phyl
- thin to med bndl. - slightly arg.
- Sulf. 25% - cg Py 20% as
diss. grains along foln.
- 5% f. As diss along foln.

M.S. - As 60%, Py 5%, g2t2 + minor
ser 34%
- ZnS 1% - v fg and diss
- As fg to mg in 5cm wide bndl
parallel to foln.
- Py - fg 1cm bndl parallel to foln
mainly within bottom 15 cm

QT2-SER Phyl - thin to med bndl.
- 15% sulf - thin to 1cm wide bndl of mono.
- mg. Py bands parallel to foln
- fine red ZnS 3% parallel to foln throughout
unit
- bottom 50 cm slightly shrd.

M.S. - 60% Sulf, 40% g2t2 and tr ser
- sulf. arc fg in mono. bndl, 1 to 2cm
wide parallel to foln
- As 48%, Py 10%, ZnS 2%
- ZnS in very fine red-brown bndl parallel
to foln mainly in lower 10 cm
- Py thin mono bndl parallel to foln mainly
in top 15 cm
- As v fg, 1 to 2cm bndl parallel to
foln.
- QT2 + minor ser in thin bndl which pinch
and swell along foln.

Lst - grey to bk - contorted med bndl
- roughly parallel with foln
- argl., tr gr on foln.
- very minor small cb sweatouts with
cg red ZnS and minor tetr.

10610.85 E

9941.116 N

PLAN

BP/SELCO

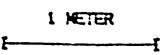
J&L PROJECT

830 LEVEL DRIFT

FACE 124

10610.85 EAST

SCALE 1:50

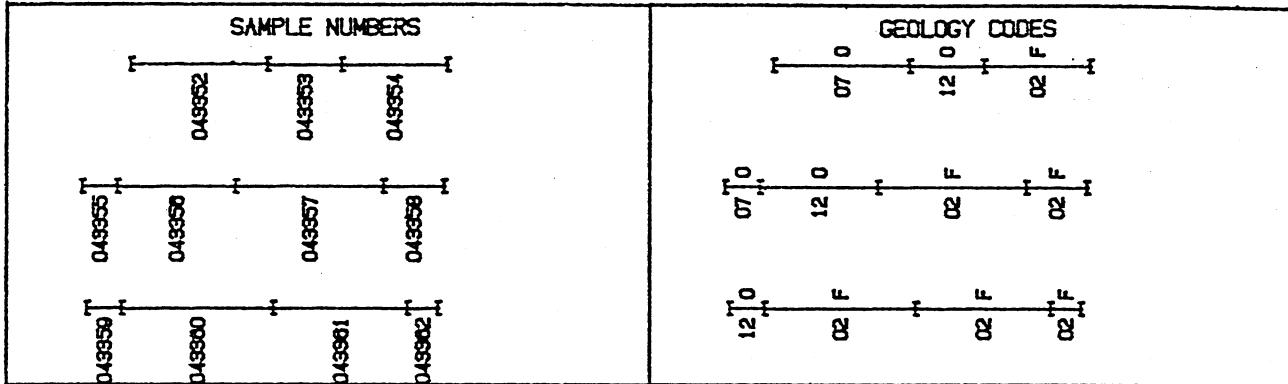


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 125

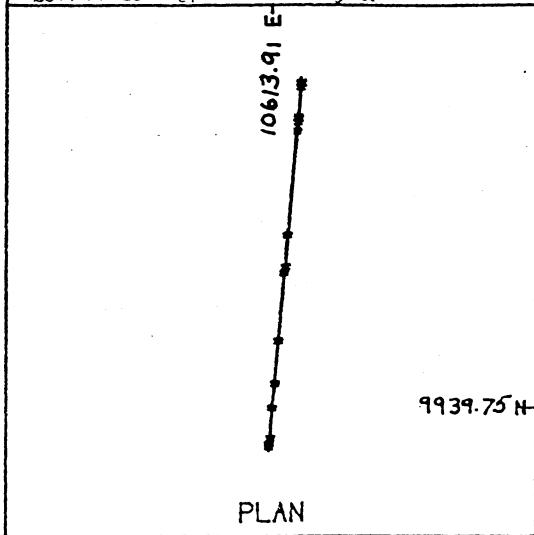
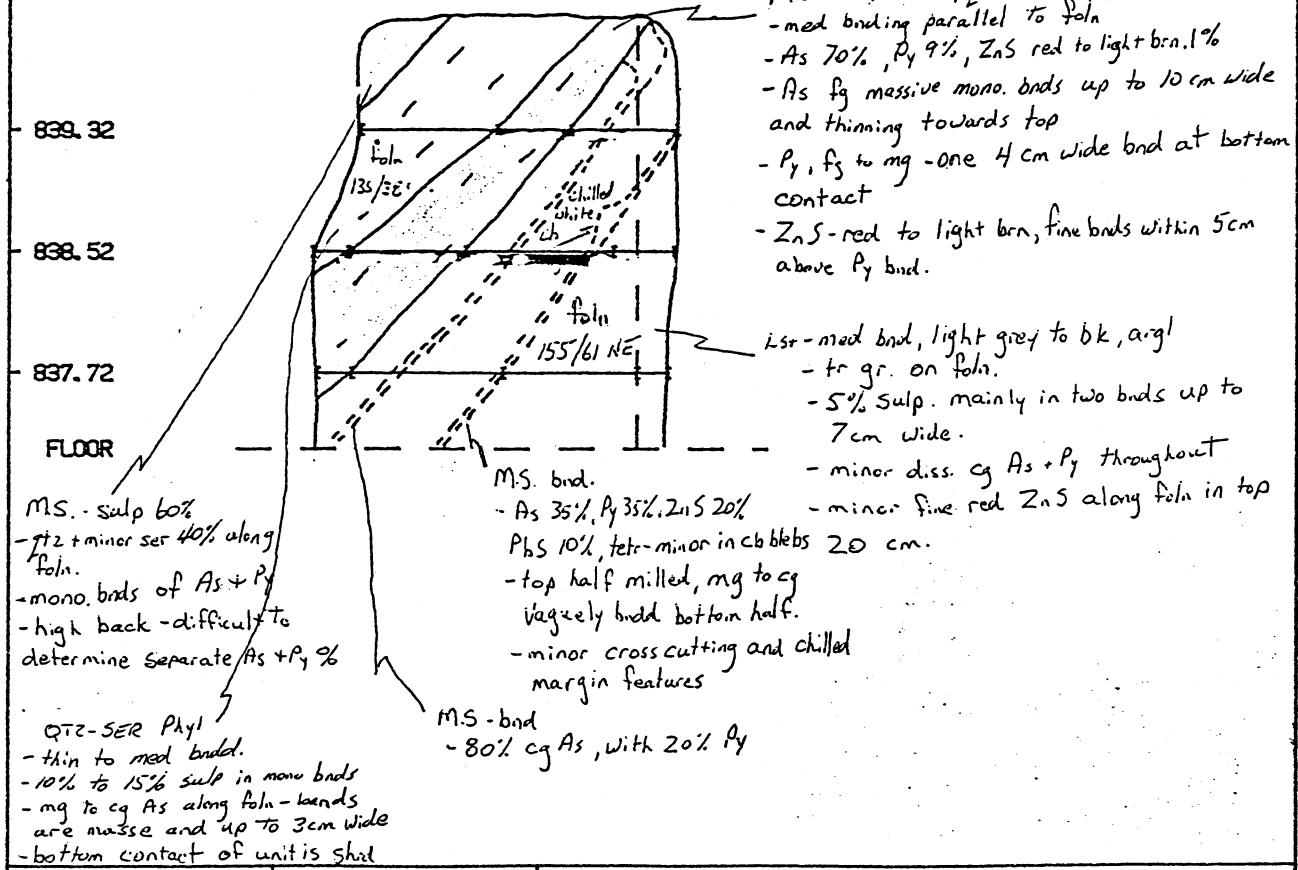
* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43352	.90	.17	.04	7.700	1.7	2.5
43353	.49	1.43	1.53	12.500	40.5	8.4
43354	.70	21.60	11.80	3.880	648.6	17.1
43355	.23	.33	.07	7.650	8.6	1.6
43356	.78	.57	.38	14.300	16.5	12.4
43357	.98	7.48	2.96	4.790	266.7	6.7
43358	.40	.18	.35	2.170	.3	2.4



GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 125

10613.91 EAST

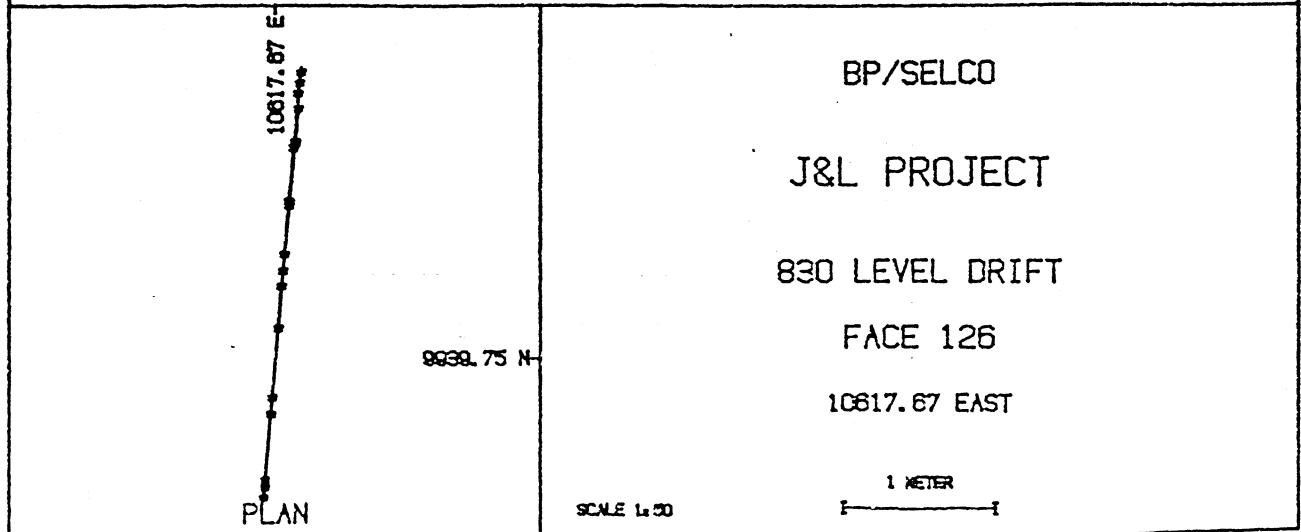
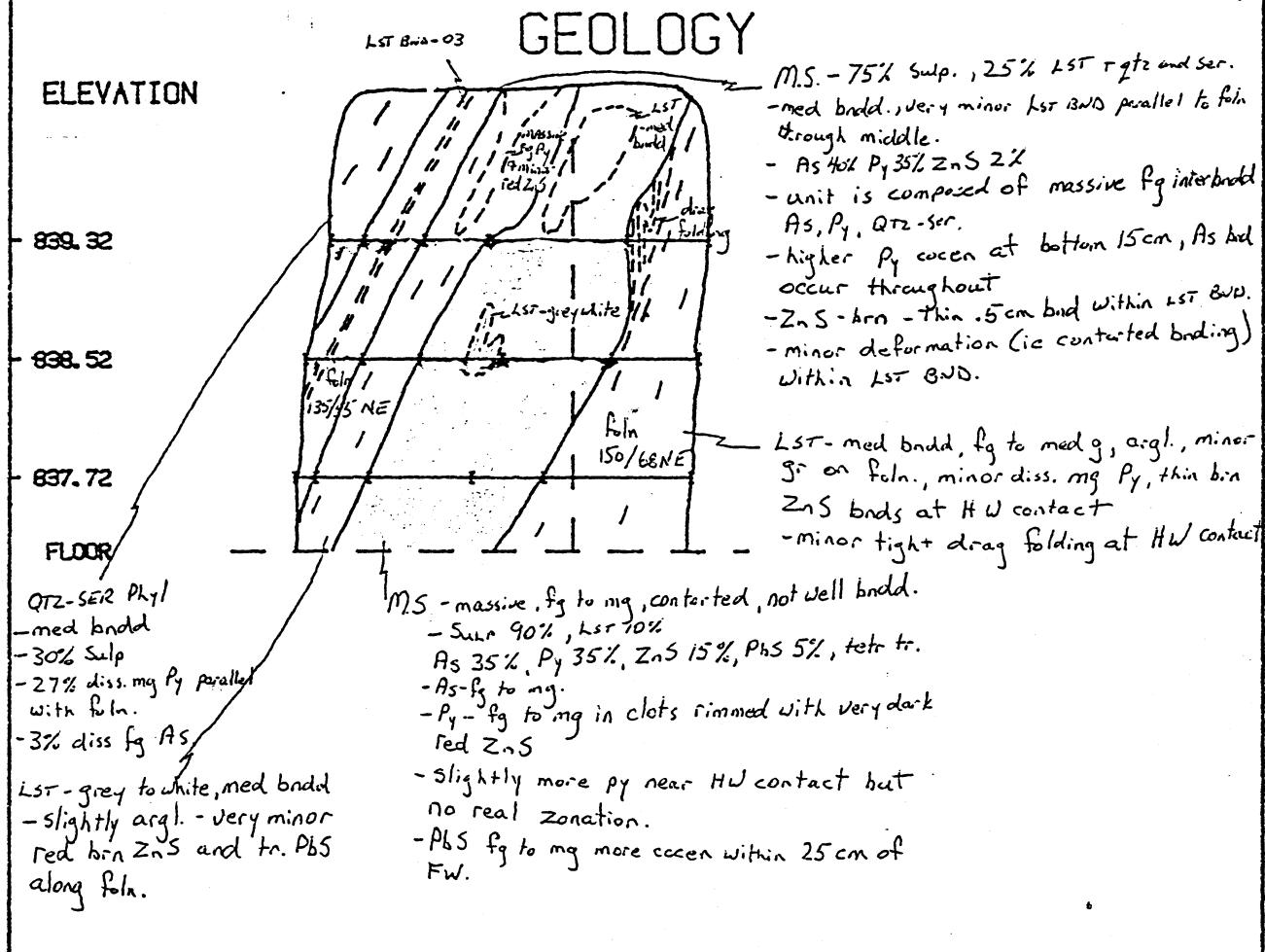
1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 126

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43363	.23	.22	.05	4.070	5.8	4.6
43364	.41	.34	.11	11.100	8.9	11.6
43365	.43	.50	.50	1.520	6.9	3.4
43366	.95	3.68	6.41	14.200	138.5	26.1
43367	.55	1.33	.68	.467	32.9	1.3
43368	.39	.50	.52	7.110	.3	2.3
43369	.39	.10	.17	.573	.7	.8
43370	.56	.19	2.12	17.800	16.5	26.6
43371	.74	4.87	15.00	9.210	136.4	24.0
43372	.59	.23	1.01	1.280	8.6	1.7

SAMPLE NUMBERS						GEOLOGY CODES					
043969	043964	043965	043966	043967		07	0	12	0	12	0
043968	043969	043970	043971	043972		12	0	0	12	0	F
043973	043974	043975	043976	043977		12	0	0	12	0	F

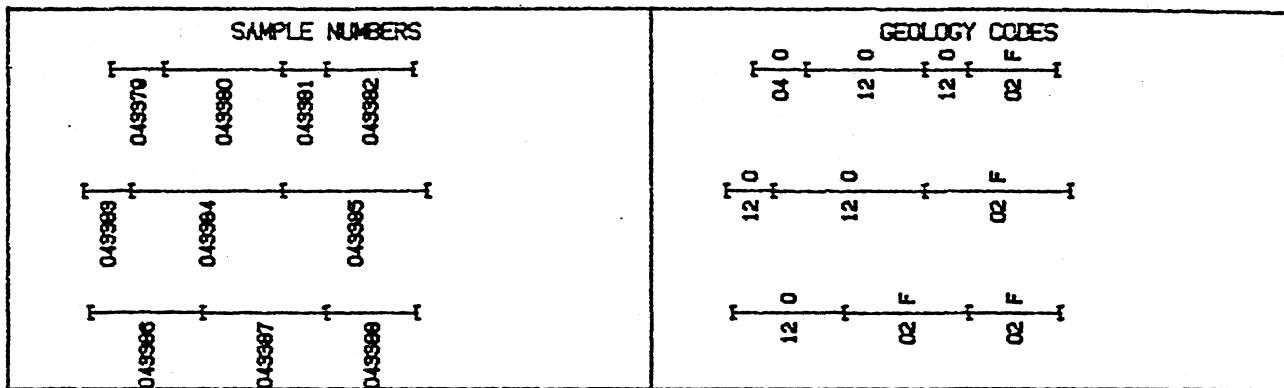


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

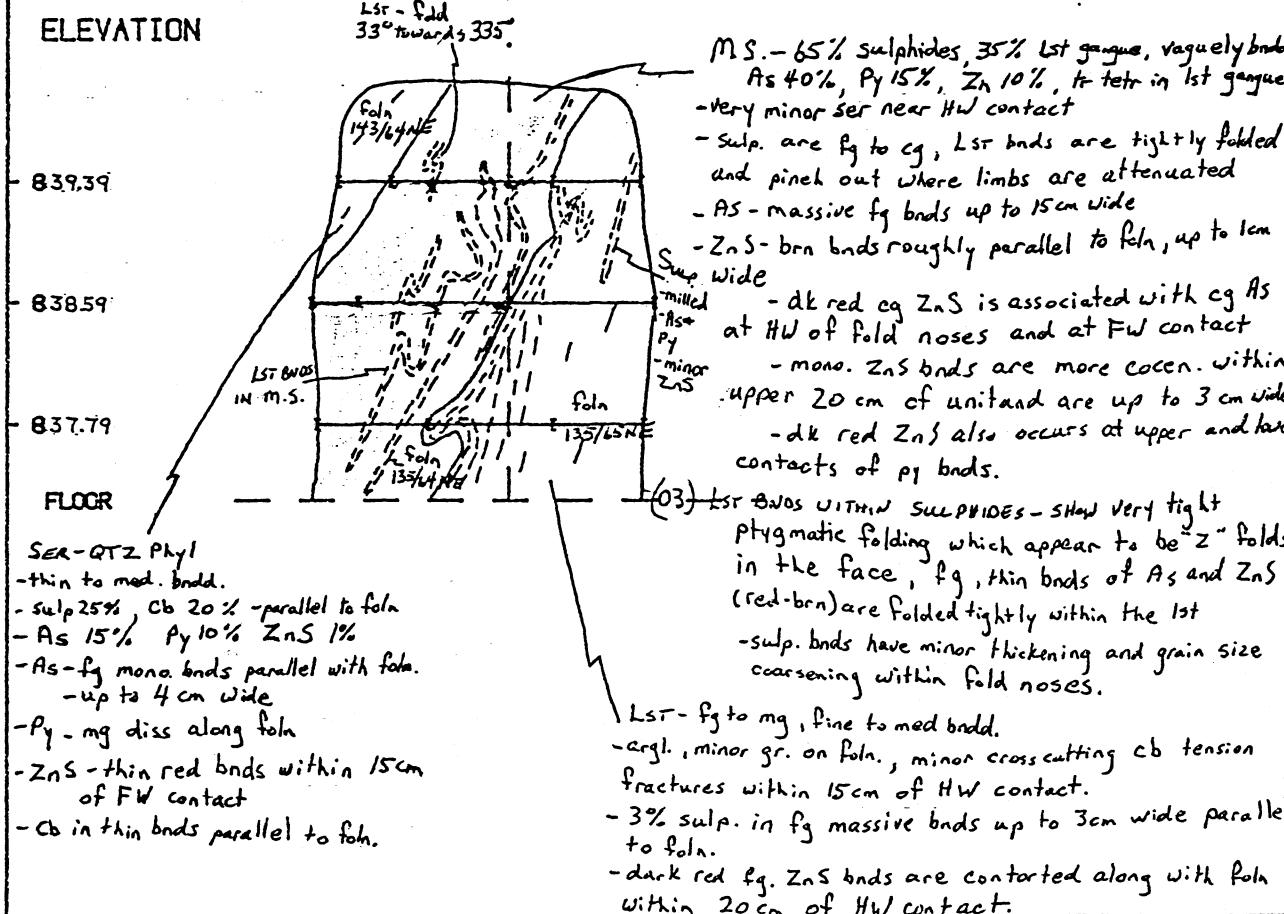
FACE# 127

* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43379	.35	.54	1.92	7.420	18.2	4.3
43380	.78	1.63	3.47	11.500	55.9	20.3
43381	.29	2.56	5.22	9.330	57.2	16.1
43382	.58	.56	.94	1.210	11.7	1.9
43383	.31	1.63	1.92	9.920	64.1	7.1
43384	1.00	1.30	1.86	7.740	48.7	10.4
43385	.76	.48	.65	1.560	13.7	2.6



GEOLOGY



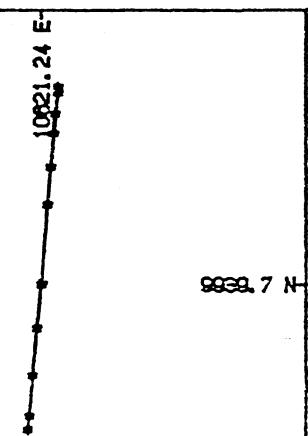
BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 127

10621.24 EAST



SCALE 1:50

1 METER

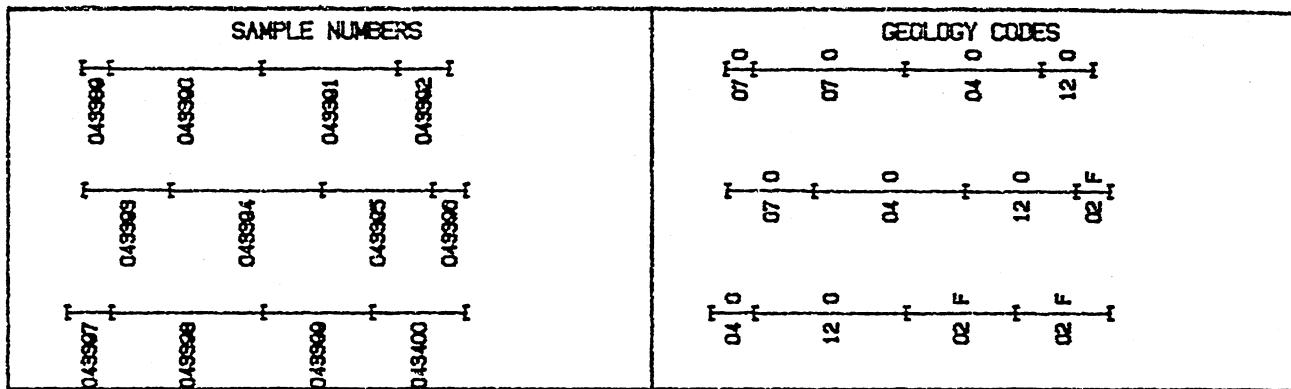
SELCO DIV. - BP RESOURCES

J & L PROJECT, B.C.

FACE# 12B

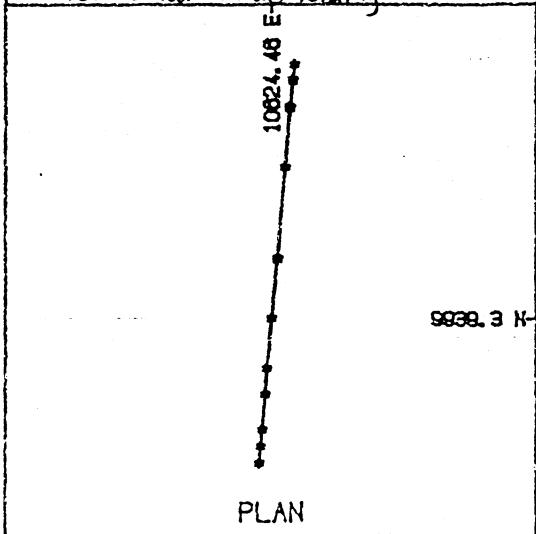
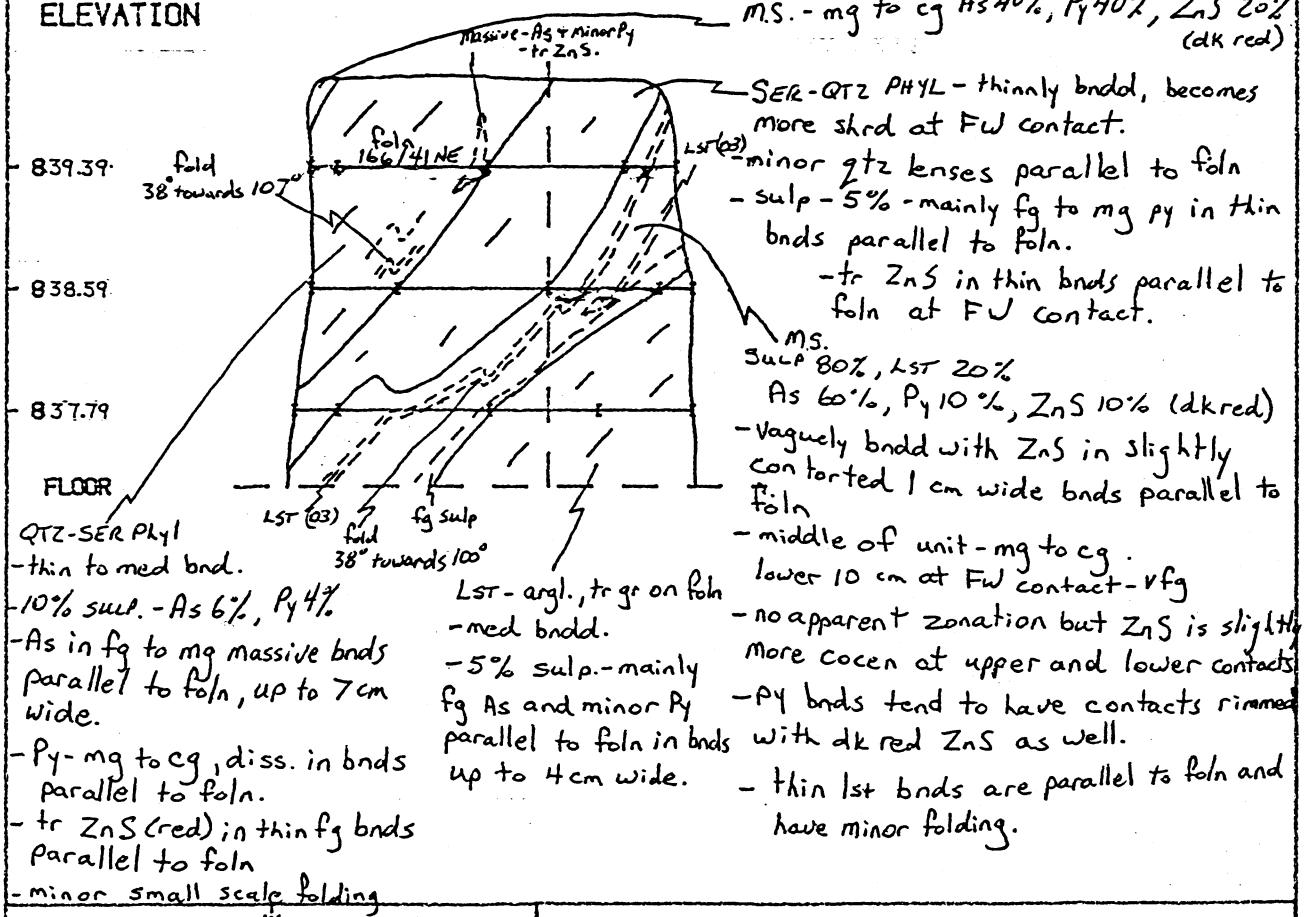
* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43389	.18	.15	.04	6.120	6.9	8.0
43390	1.00	.18	.03	2.940	4.1	4.7
43391	.90	.29	.07	2.200	6.2	1.4
43392	.34	.70	4.85	19.100	16.1	33.4
43393	.57	.27	.06	1.360	5.1	1.3
43394	1.00	.29	.23	1.510	12.3	3.0
43395	.73	13.70	15.00	7.160	227.7	17.0
43396	.22	.42	.54	.623	.3	1.3



GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 128

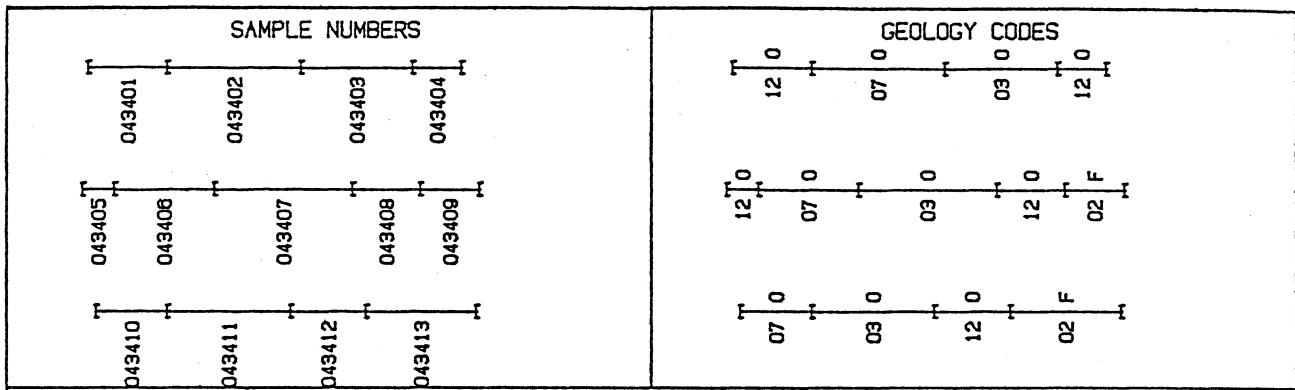
10624.46 EAST

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 129

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (o/mT)	GOLD (o/mT)
43401	.52	1.81	8.69	15.800	69.6	37.0
43402	.88	.43	.22	1.560	15.1	1.7
43403	.74	.17	.29	1.090	.3	1.5
43404	.32	12.80	13.10	8.200	249.6	18.5
43405	.21	1.53	1.81	17.100	27.8	28.2
43406	.66	.32	.09	1.490	14.1	.9
43407	.91	.26	.54	1.930	5.5	3.1
43408	.45	1.08	5.14	9.150	23.7	13.2
43409	.39	.03	.18	.234	.3	.3



GEOLOGY

ELEVATION

839.39

838.59

837.79

FLOOR

M.S. - massive, very vaguely bndd
 - mg to cg, As in lower 60 cm vgc
 As 45%, Py 40%, ZnS 15%, PbS 1%
 Cpytr, minor gte
 - Py fg to mg in 10 cm wide irregular
 bnds rimmed with very dk red ZnS
 bnds + str.
 - higher Py vgc. near FW contact
 - ZnS-dk red in .5 to 3 cm irregular
 bnds + str.
 - PbS - diss. mg to cg

milled
surf.

cg
surf.

foln

141/47NE

foln

141/175NE

Joint

154/52SW

LST - slightly argl, med bndd.
 - 2% ser in thin bnds parallel to foln, foln changes downdip
 - 4% sulp. - 3% fg lsm As bnds with thin brn ZnS bnds parallel to foln throughout unit.

- Py - fg to mg, mainly in lower 15 cm, massive to vaguely bndd.
 - ZnS - fg, red to red-brn mainly in lower half.

LST - argl, minor gr on foln., med bndd.
 - few tension fractures with white cb at HW contact.
 - minor diss. fg py.

10628.03 E

10628.03 N

9938.79 N

PLAN

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 129

10628.03 EAST

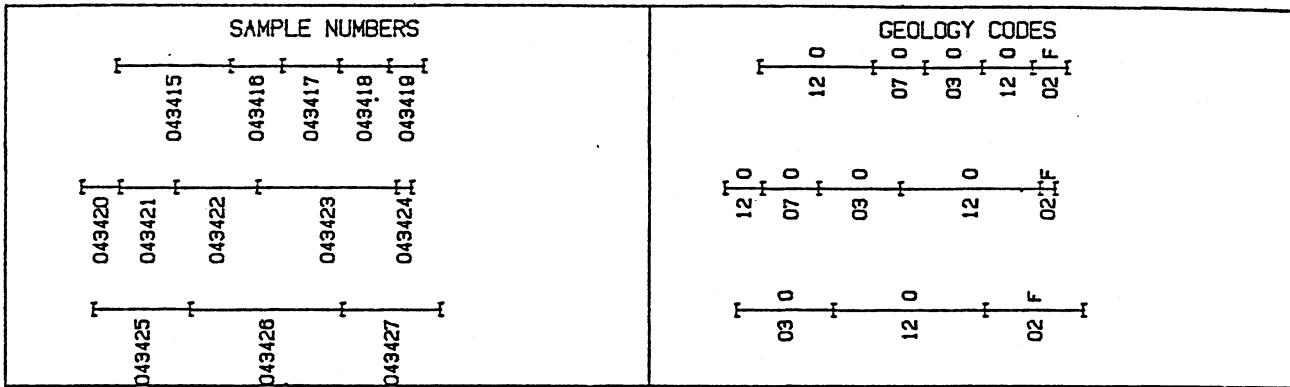
SCALE 1:50

1 METER

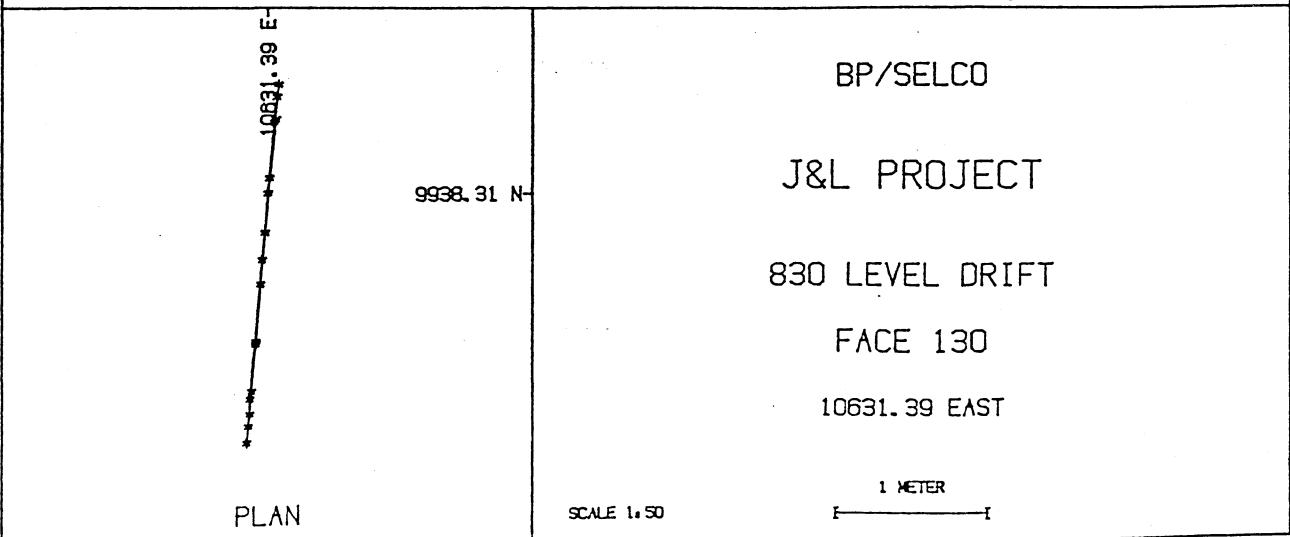
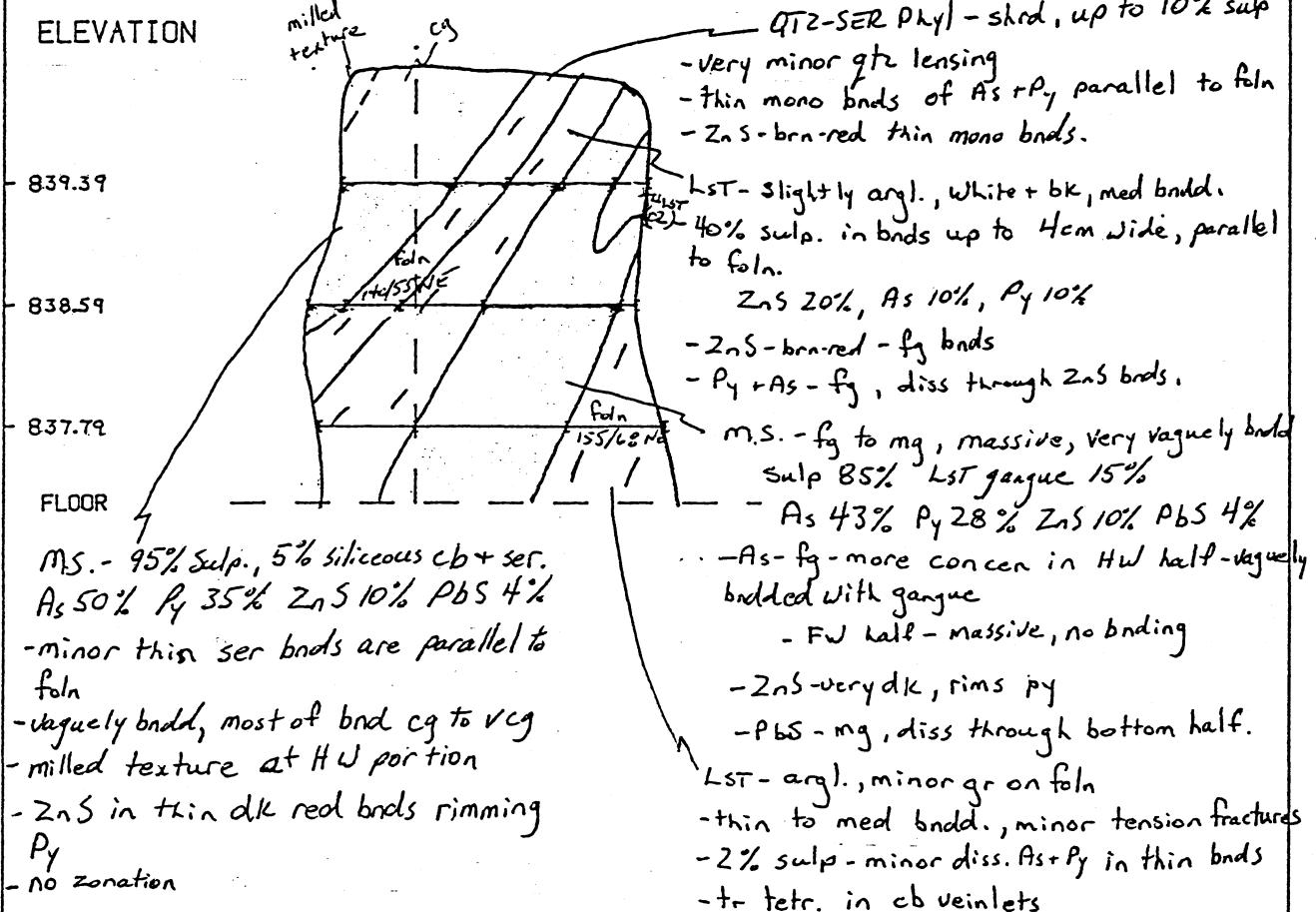
SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 130

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43415	.75	3.23	5.06	10.900	81.6	17.1
43416	.34	1.08	1.30	4.180	46.6	9.1
43417	.38	1.54	3.05	3.510	31.9	7.7
43418	.33	.06	.05	14.000	3.8	14.2
43419	.23	.09	.13	3.450	4.8	4.4
43420	.25	4.28	11.50	7.010	209.5	18.2
43421	.37	.58	.38	2.670	21.9	2.5
43422	.54	.90	1.62	3.480	16.8	4.8
43423	.92	3.36	4.52	15.100	45.6	26.9
43424	.10	.33	.36	1.460	.3	1.7



GEOLOGY

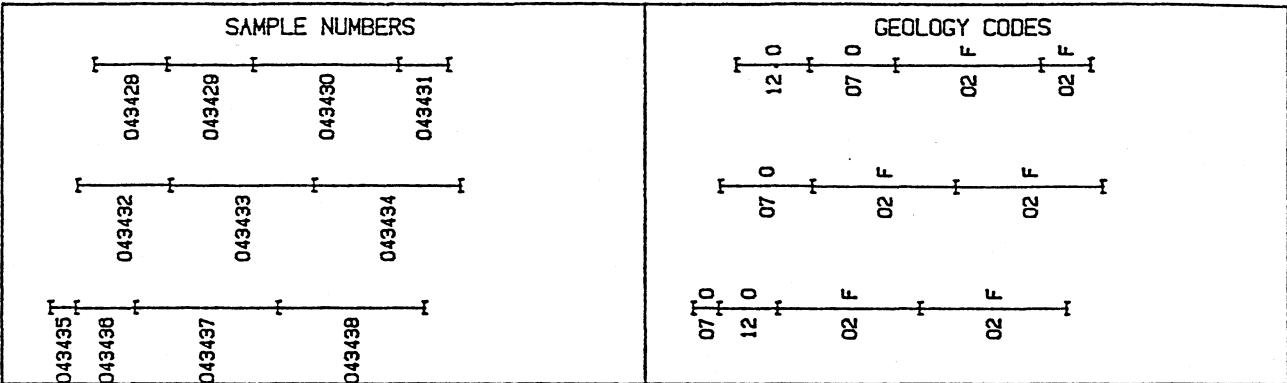


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

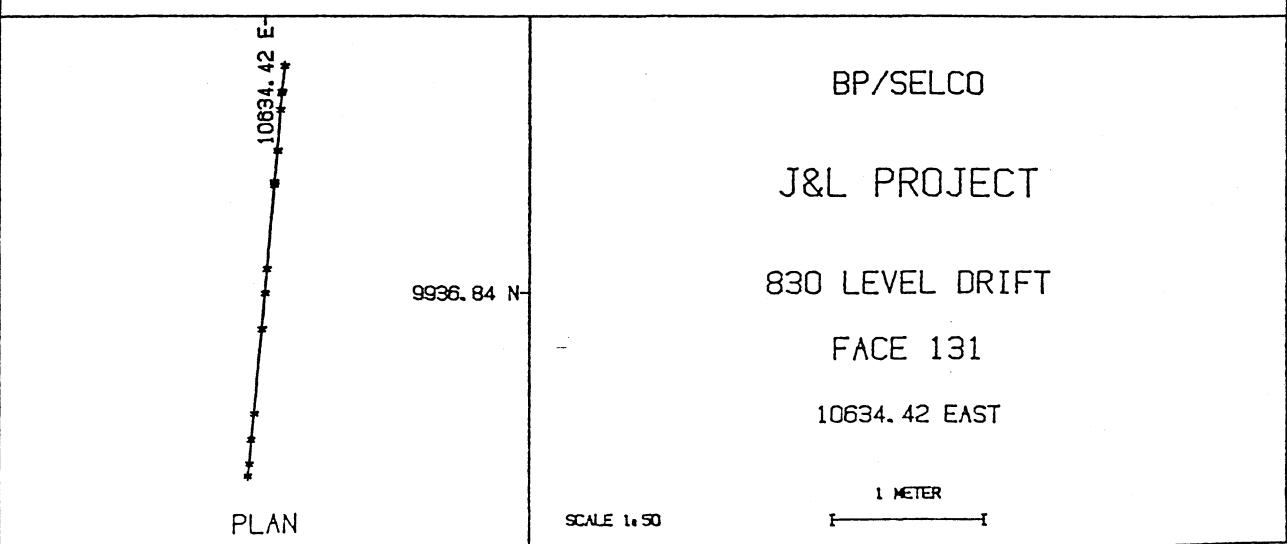
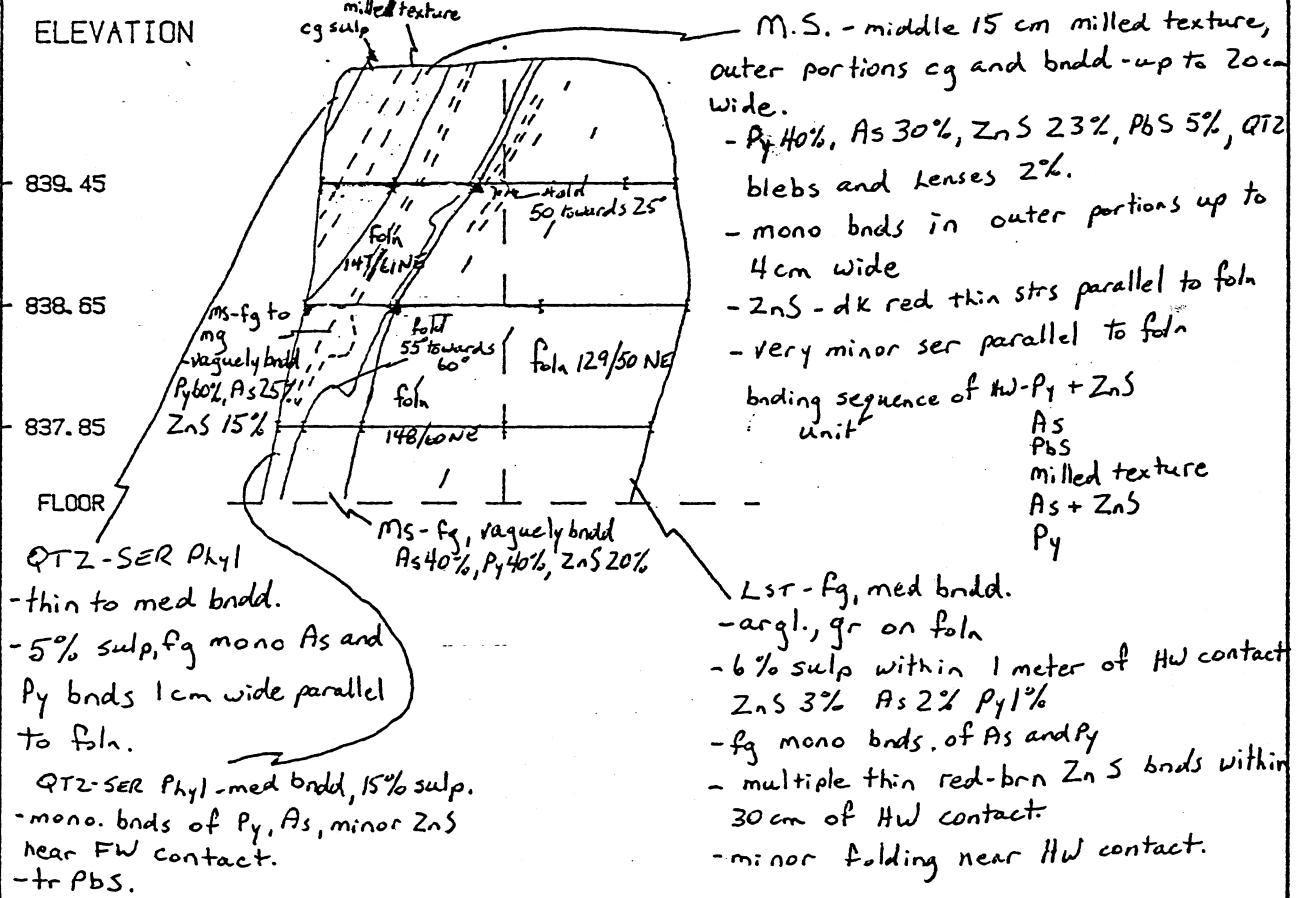
FACE# 131

* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43428	.48	7.46	16.60	8.290	150.2	25.5
43429	.57	2.42	4.79	7.200	74.4	8.8
43430	.96	.46	.33	1.410	10.6	3.8
43431	.33	.04	.03	.916	.3	.7
43432	.61	1.03	1.67	5.860	45.3	5.3
43433	.95	1.00	.72	5.640	22.3	10.6
43434	.97	.04	.07	.301	.3	.4



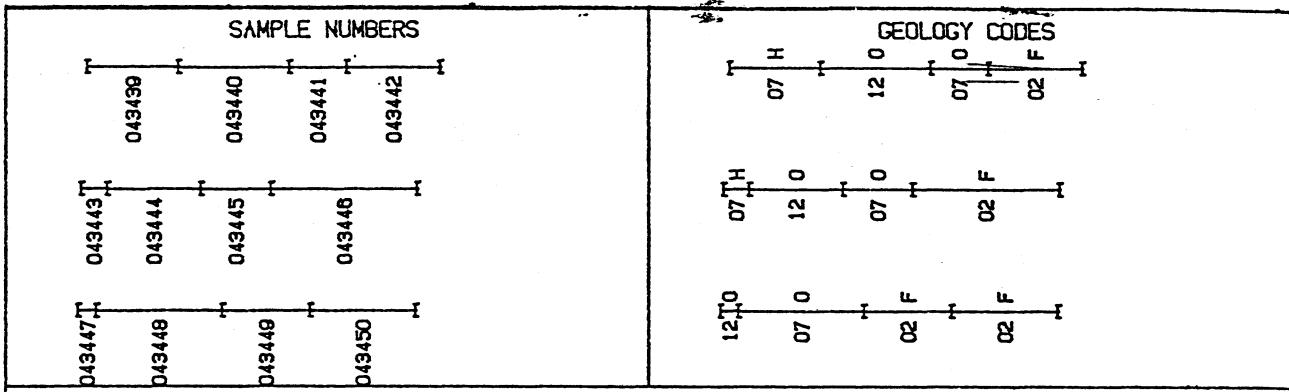
GEOLOGY



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

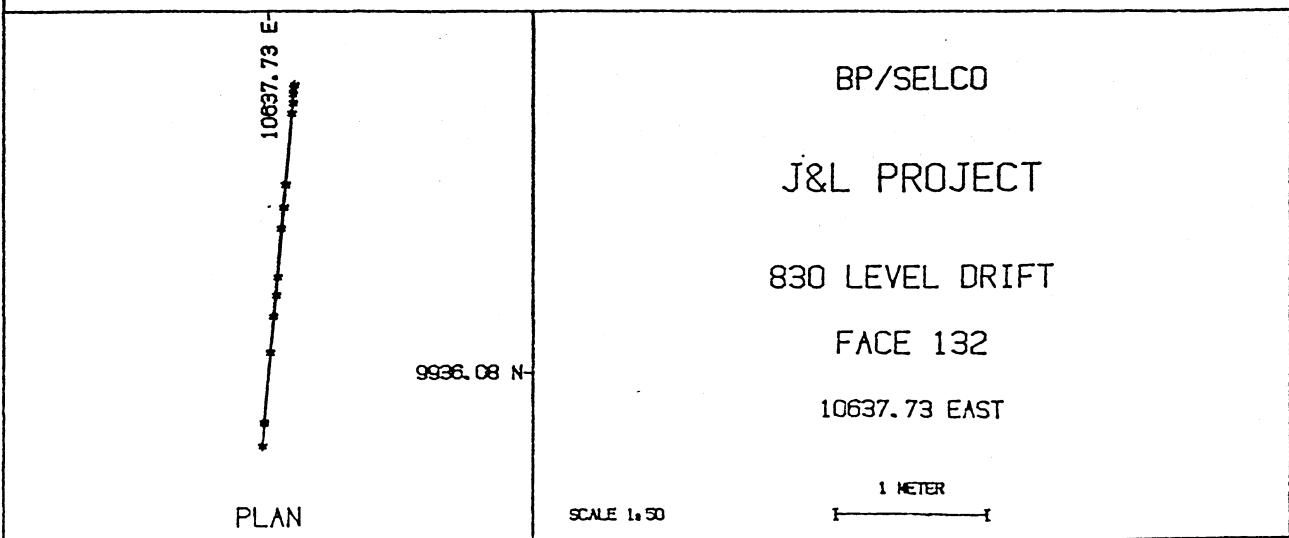
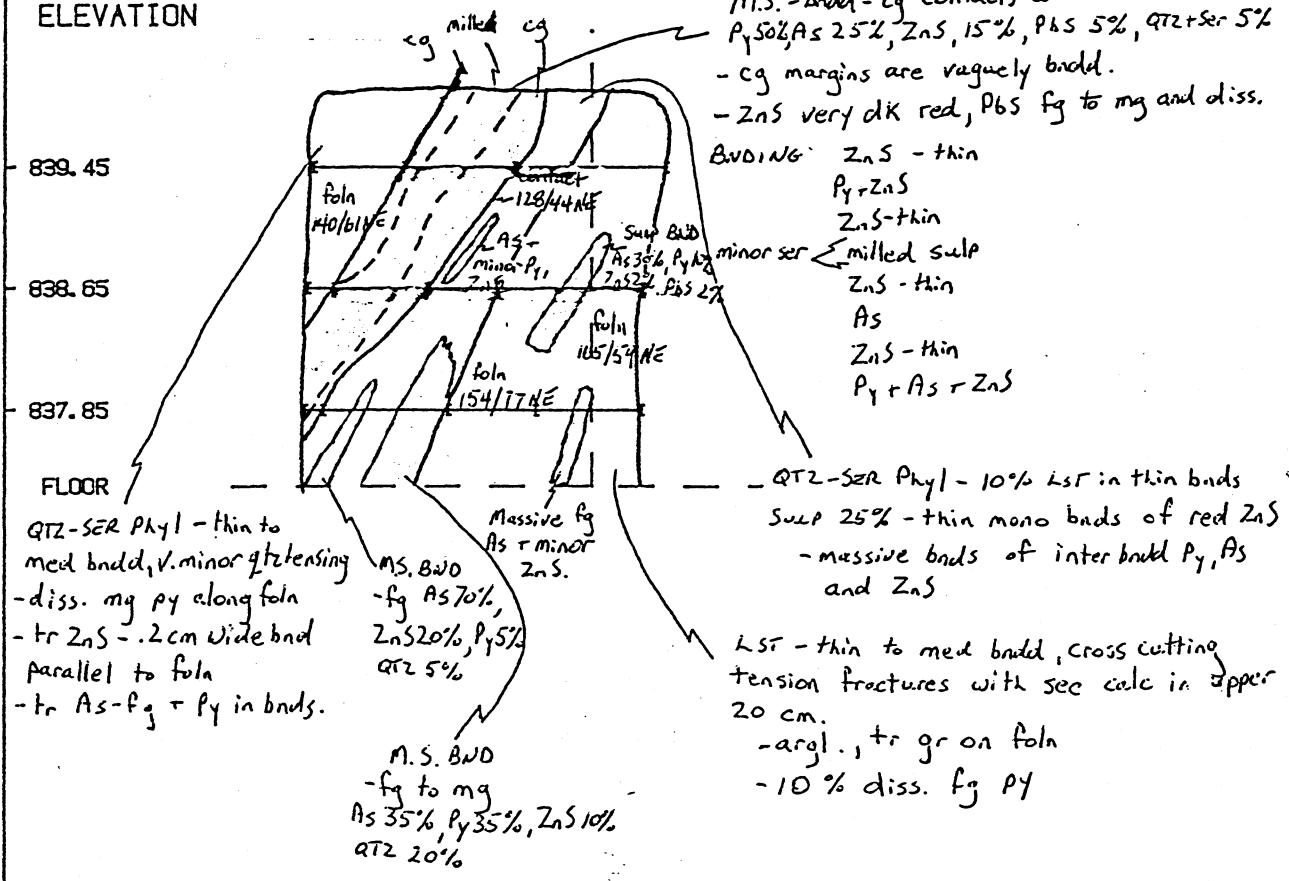
FACE# 132

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43439	.60	.03	.07	.372	3.1	.3
43440	.73	2.45	10.90	8.050	177.3	20.0
43441	.38	1.39	1.13	10.300	41.5	11.7
43442	.62	.55	.43	1.980	11.0	2.5
43443	.17	.09	.06	.617	7.2	.8
43444	.62	6.15	15.30	8.520	185.1	19.5
43445	.46	2.90	3.58	8.190	104.9	13.0
43446	.97	.81	.42	6.740	75.8	22.6



GEOLOGY

ELEVATION

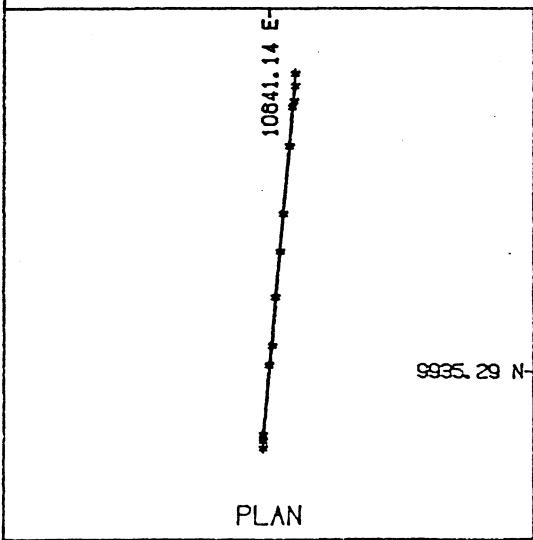
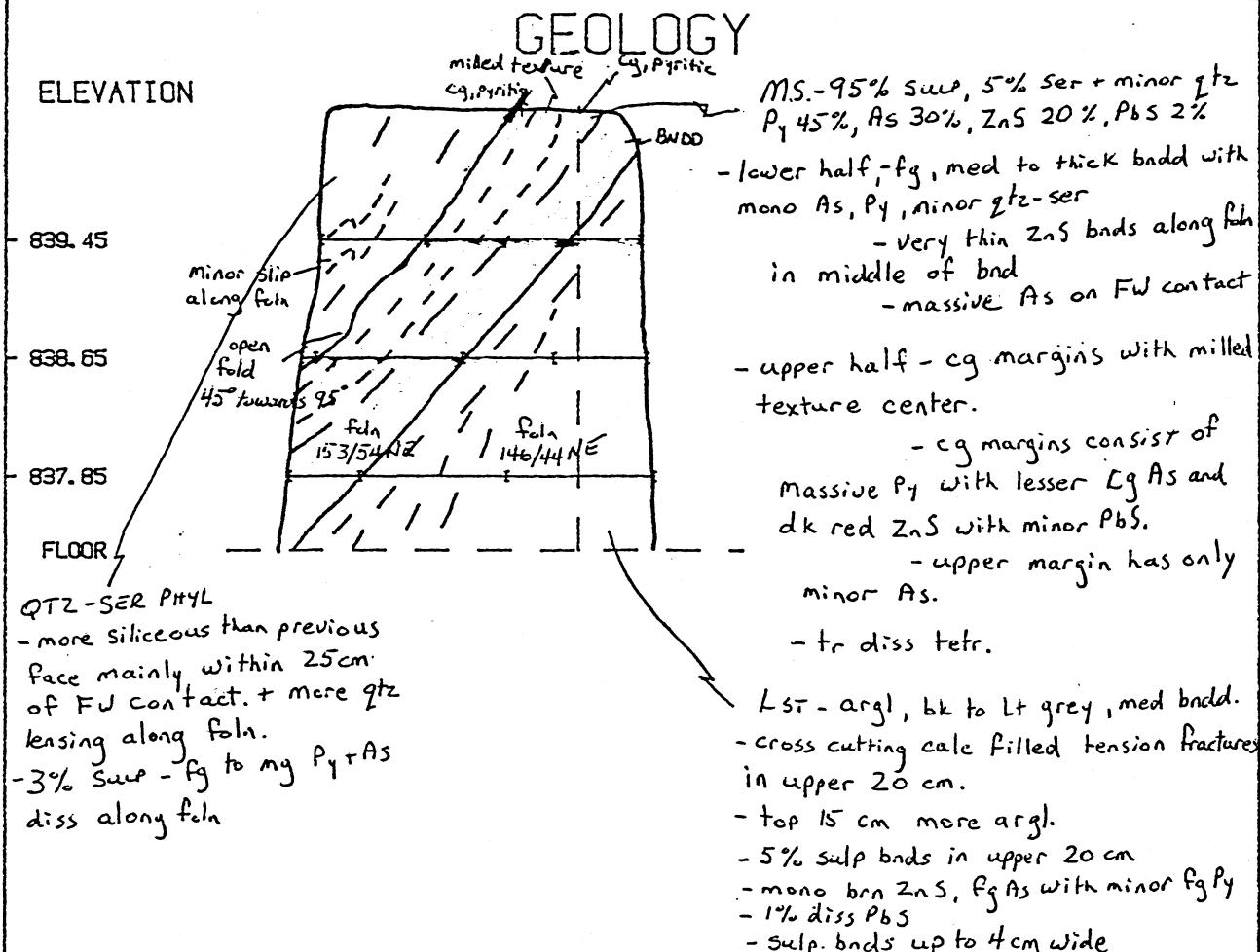
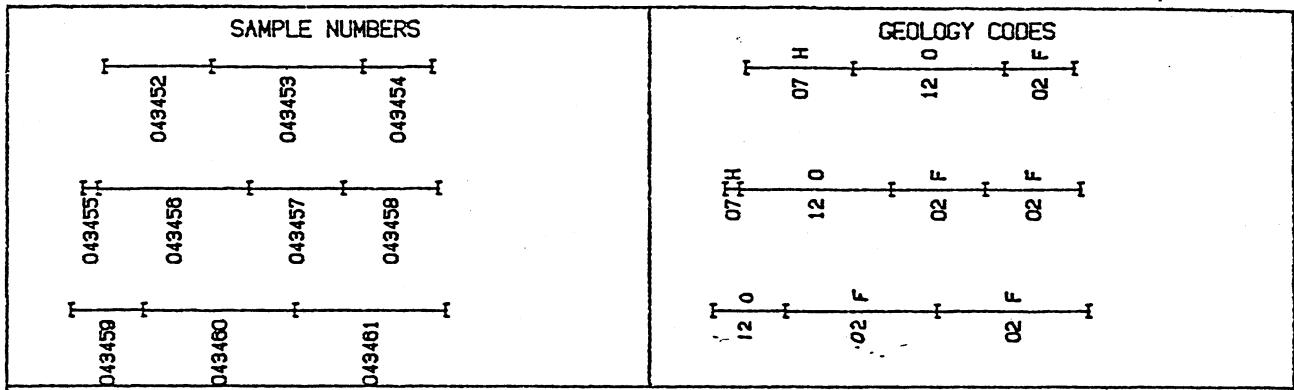


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 133

■■■■■■■■■■■■■■■■■■

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43452	.71	.03	.03	.623	4.8	.5
43453	1.00	3.86	6.72	6.820	118.6	14.7
43454	.46	3.44	.60	3.600	74.1	6.1
43455	.10	.06	.02	.438	26.4	.2
43456	1.00	4.59	7.44	13.400	215.3	25.3
43457	.62	.30	.42	1.500	.3	1.8
43458	.63	.06	.15	.069	.3	.2



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 133

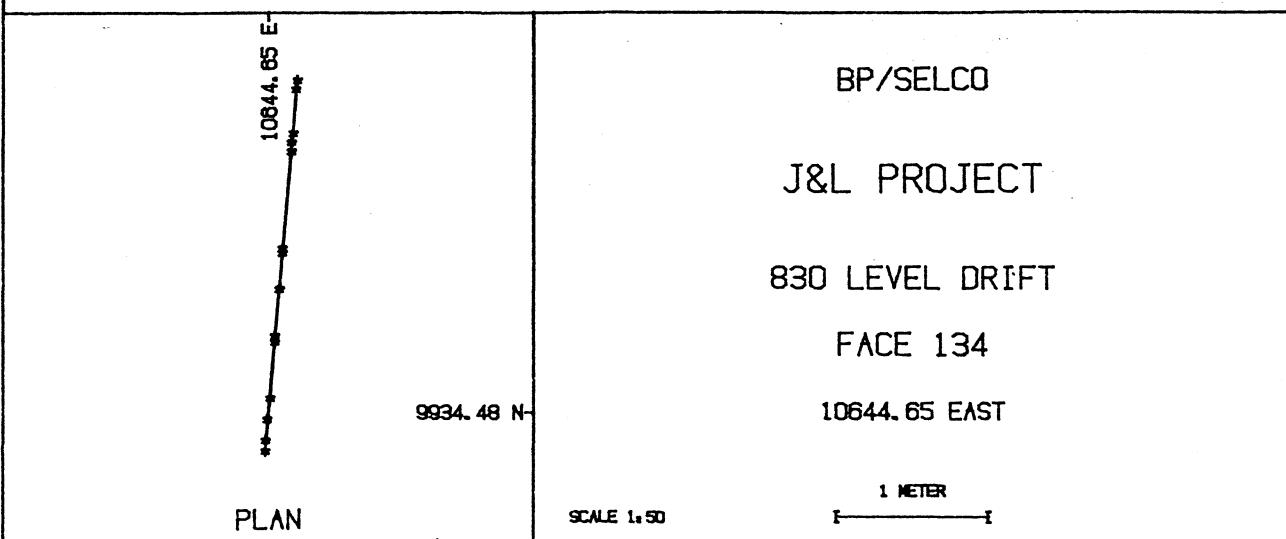
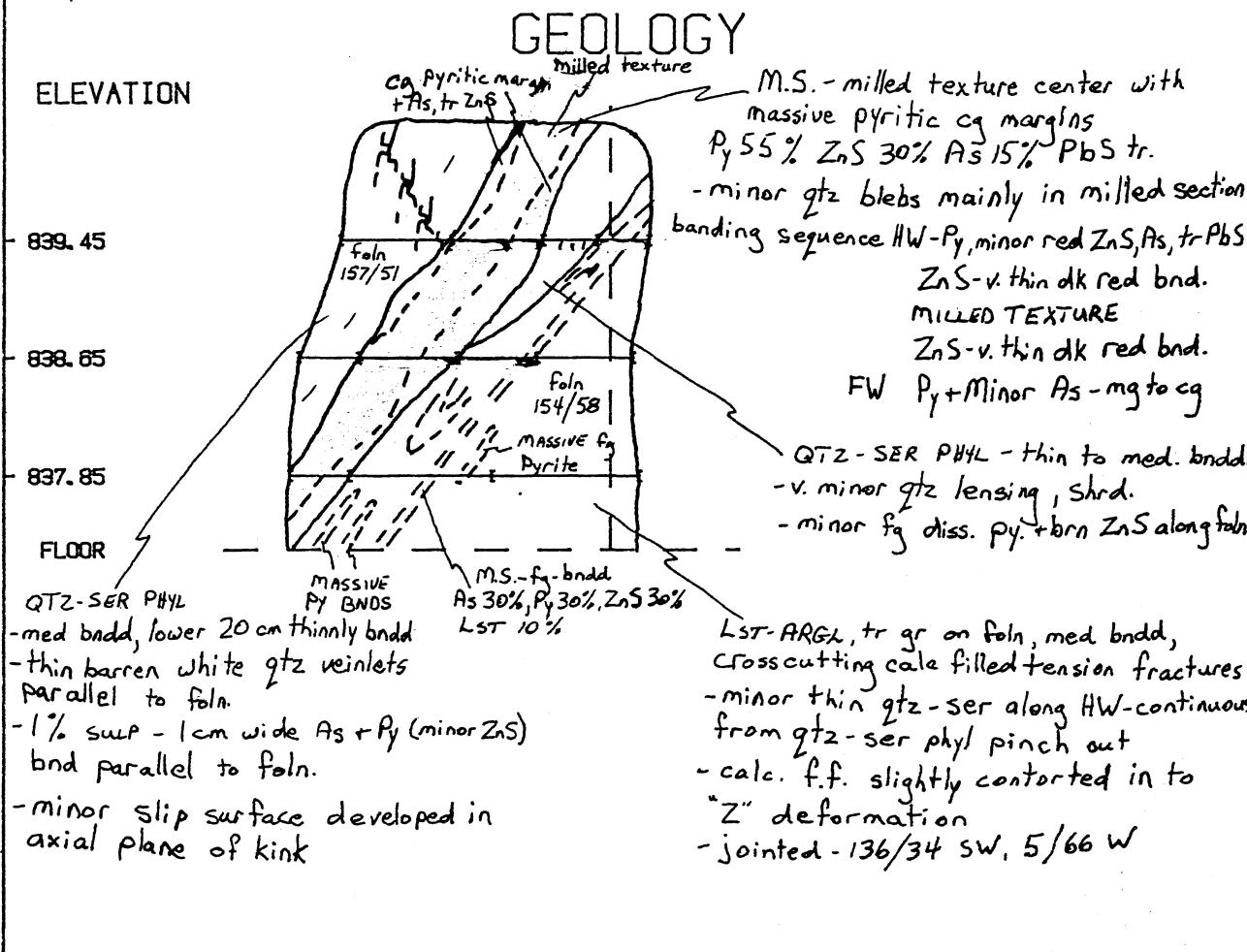
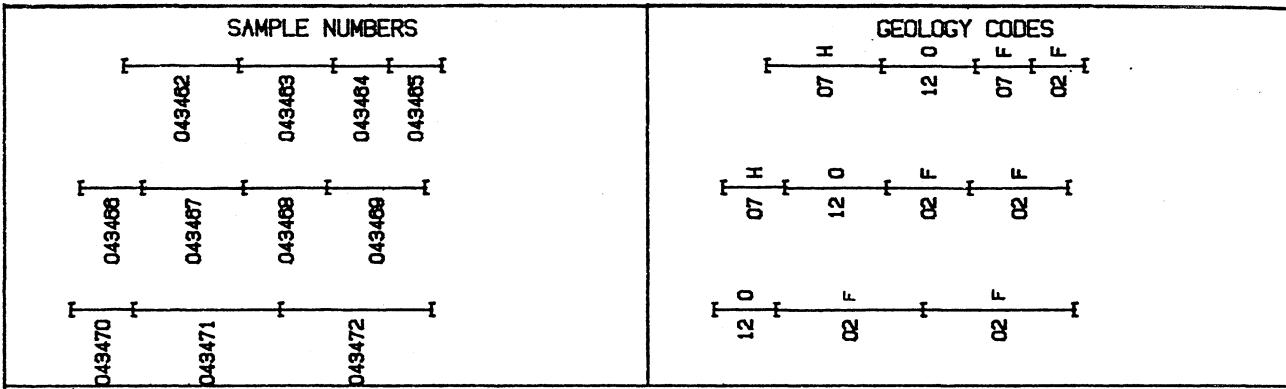
10641.14 EAST

SELCO DIV. - BP RESOURCES

J & L PROJECT, B.C.

FACE# 134

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43462	.76	.13	.14	.439	32.6	.5
43463	.62	5.48	15.00	4.570	191.3	15.9
43464	.37	.71	.45	2.550	40.1	2.3
43465	.35	2.74	3.10	3.040	84.3	9.0
43466	.41	.08	.18	.978	12.3	.7
43467	.67	4.62	11.10	4.210	312.3	18.1
43468	.55	.55	1.43	3.020	43.5	5.1
43469	.65	.10	.13	.338	6.8	.5

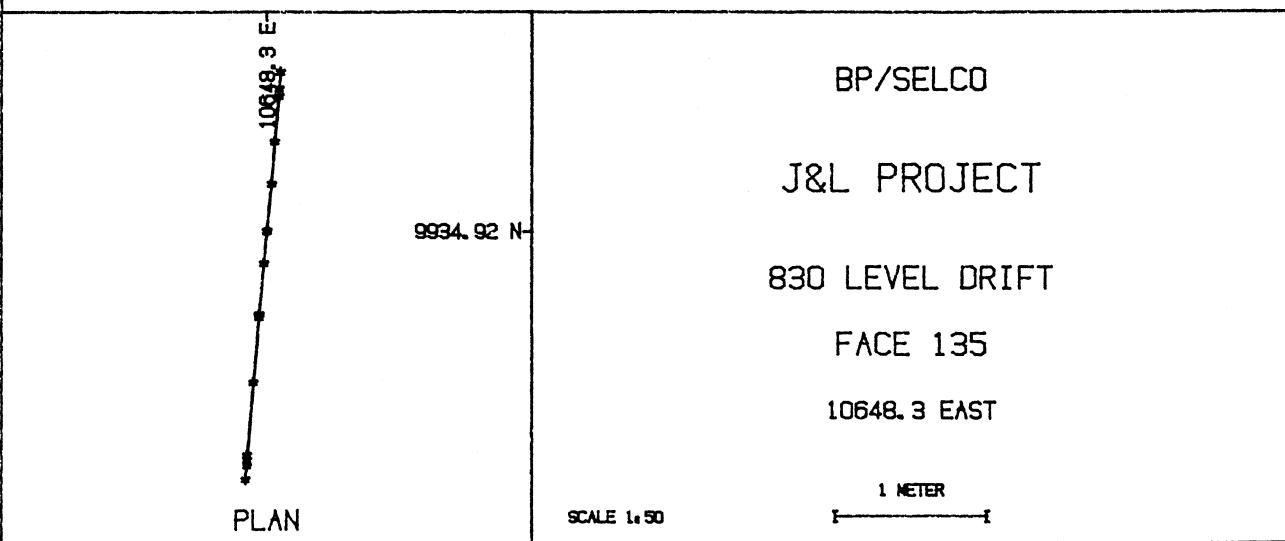
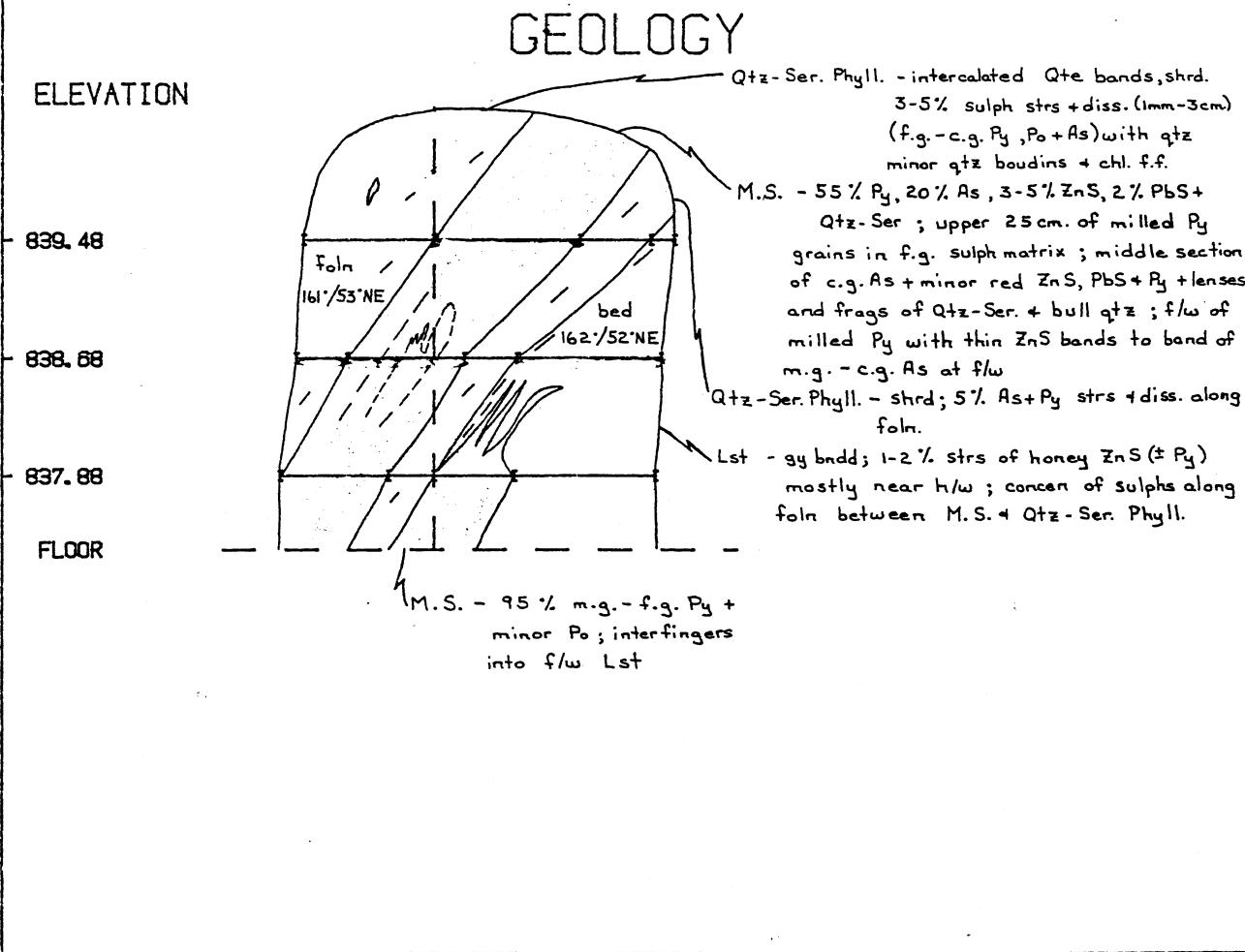
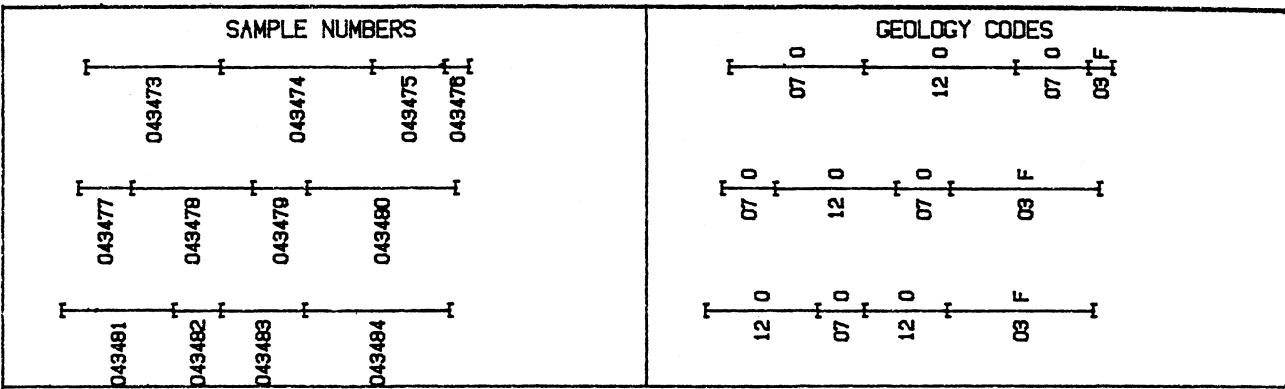


SELCO DIV. - DP RESOURCES J & L PROJECT, B.C.

FACE# 135

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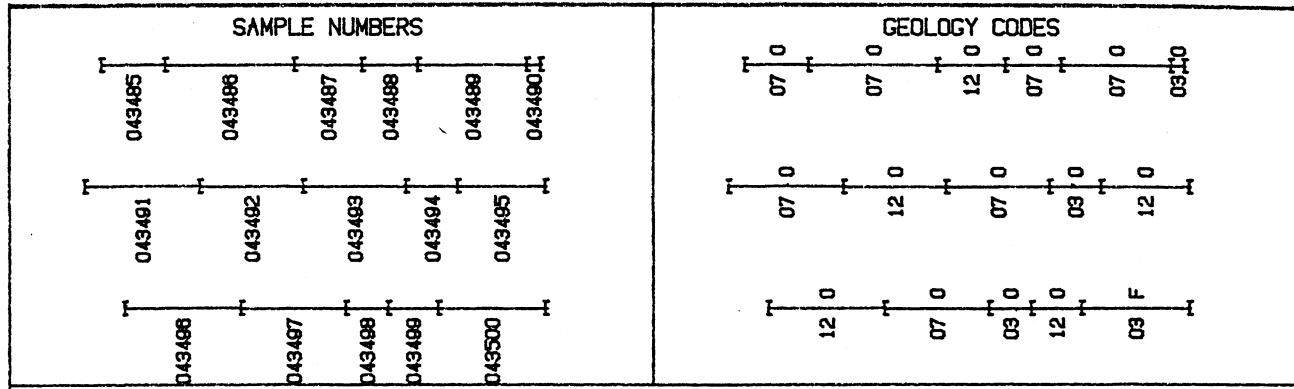
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43473	.89	.06	.04	.808	8.9	.6
43474	1.00	4.74	10.30	13.500	117.6	27.6
43475	.48	.10	.13	.723	7.9	.9
43476	.16	.10	.09	.169	7.2	.3
43477	.35	.35	1.28	1.890	17.5	5.1
43478	.80	6.49	10.70	8.100	139.2	19.1
43479	.36	.14	.13	1.650	8.2	.8
43480	.98	.44	.51	1.430	9.2	6.6



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 136

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43485	.42	.02	.05	.057	1.0	1.0
43486	.85	.02	.44	1.680	3.1	3.3
43487	.45	2.90	14.60	8.840	81.6	38.4
43488	.37	.16	.29	.897	5.5	2.6
43489	.72	.04	.13	.163	5.5	1.4
43490	.09	.45	.94	4.030	15.1	9.0
43491	.76	.04	.48	2.190	2.7	4.1
43492	.68	2.84	13.50	9.540	82.6	36.7
43493	.68	.20	.25	.909	7.9	1.8
43494	.34	.32	.51	4.760	8.9	10.4
43495	.58	.03	.03	1.480	1.4	4.9



GEOLOGY

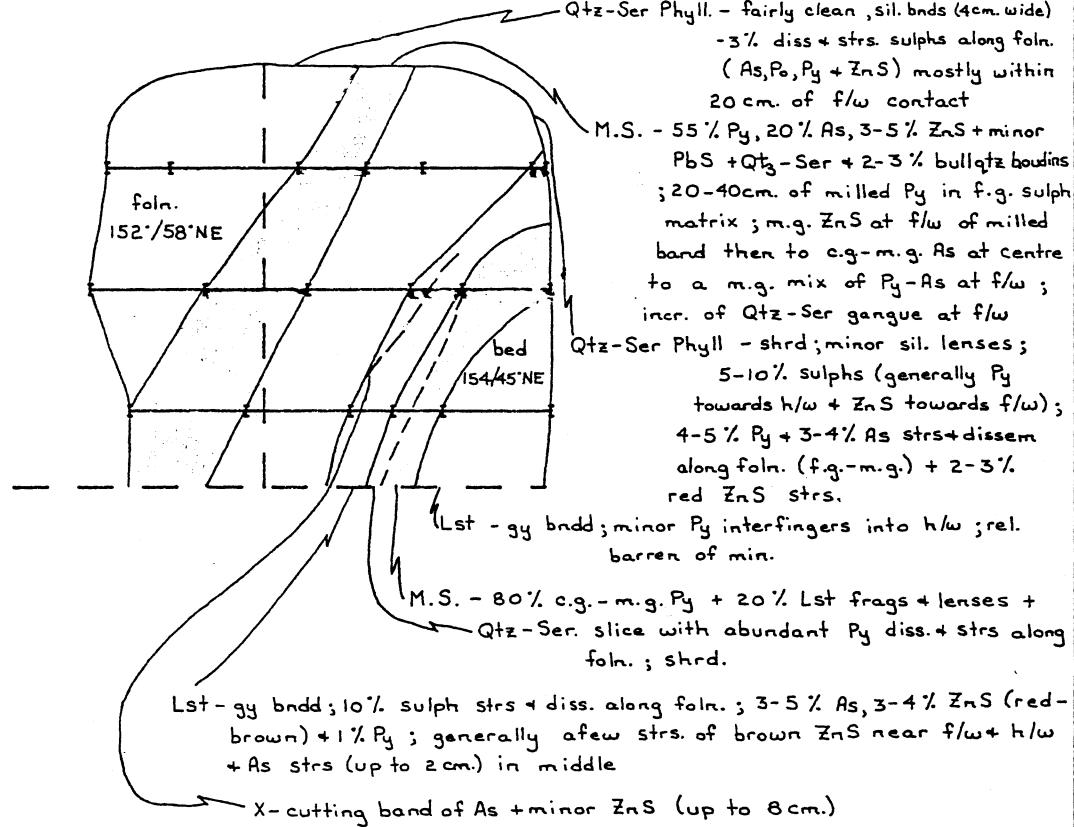
ELEVATION

839.48

838.68

837.88

FLOOR



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 136

10651.76 EAST

10651.76 E
9934.37 N
PLAN

SCALE 1:50

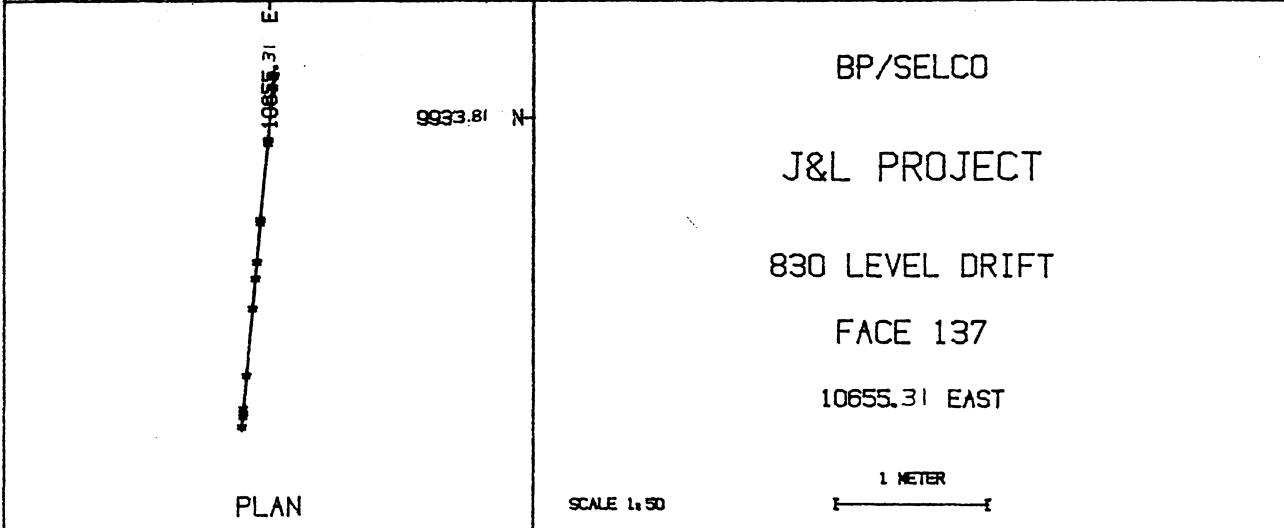
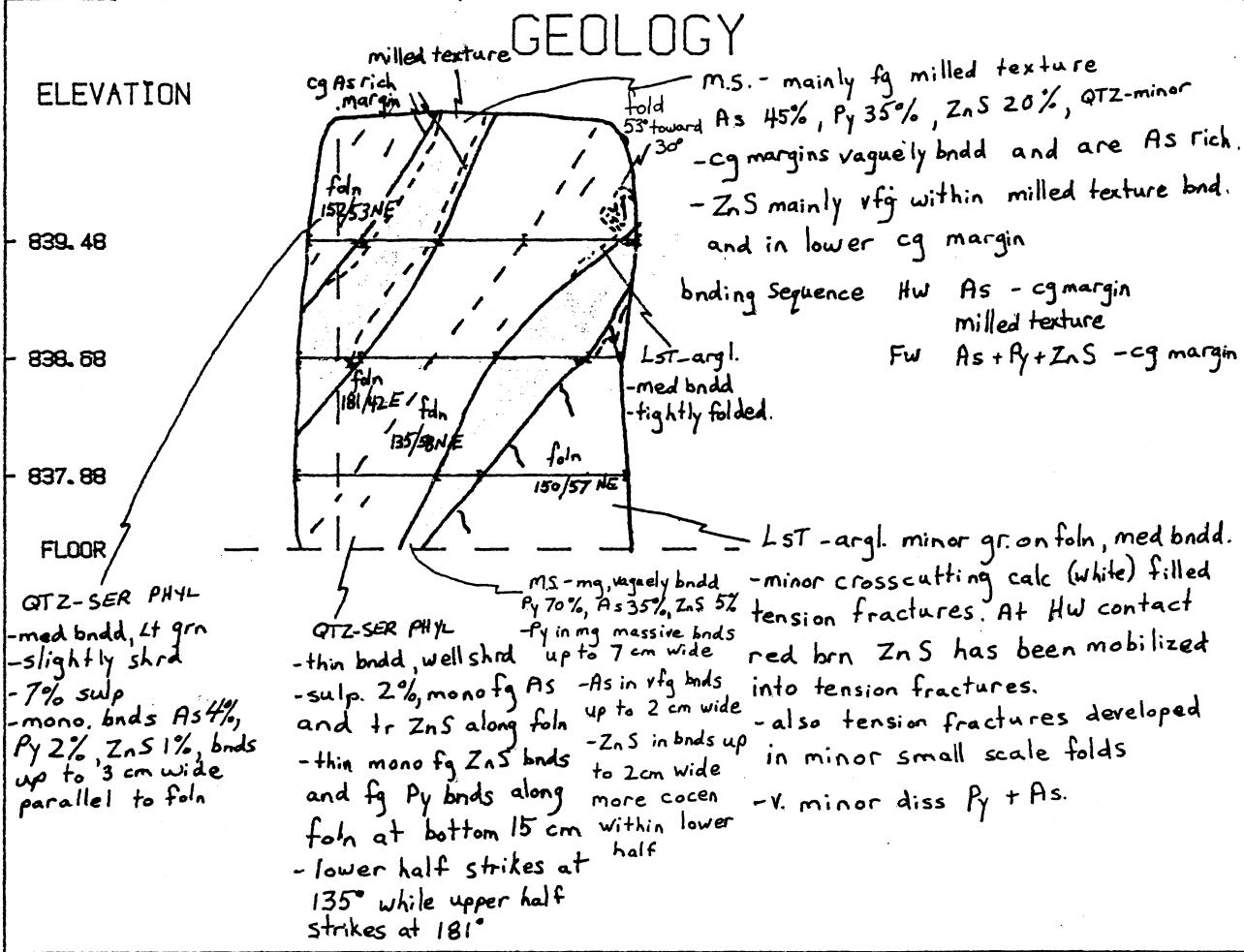
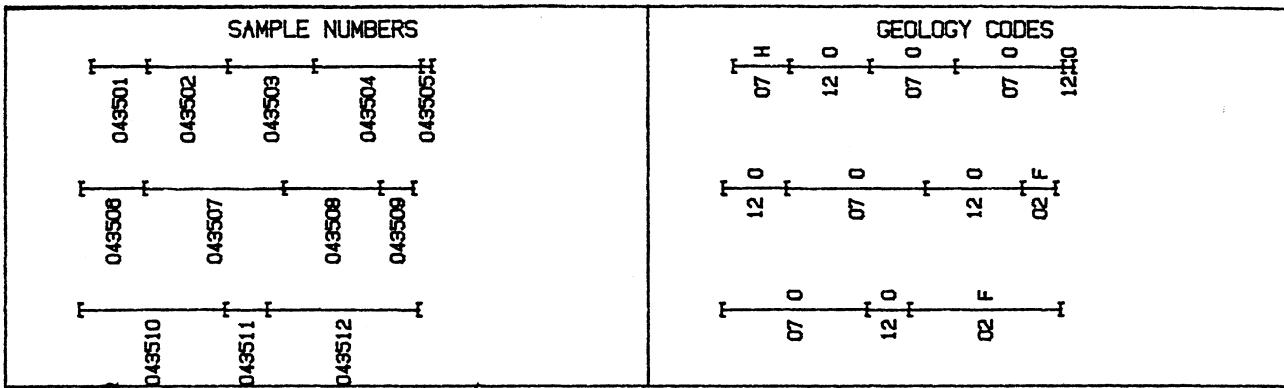
1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 137

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SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43501	.37	.01	.39	2.120	3.1	3.8
43502	.53	4.06	7.68	12.800	106.0	52.6
43503	.57	.17	.17	1.870	11.0	.8
43504	.71	.41	.74	2.880	17.5	3.8
43505	.07	3.08	1.51	7.840	82.9	19.2
43506	.42	5.69	11.80	8.900	161.1	39.8
43507	.92	.41	.52	1.900	12.7	3.1
43508	.64	.79	1.20	10.400	53.6	15.2
43509	.22	.03	.03	.124	5.1	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

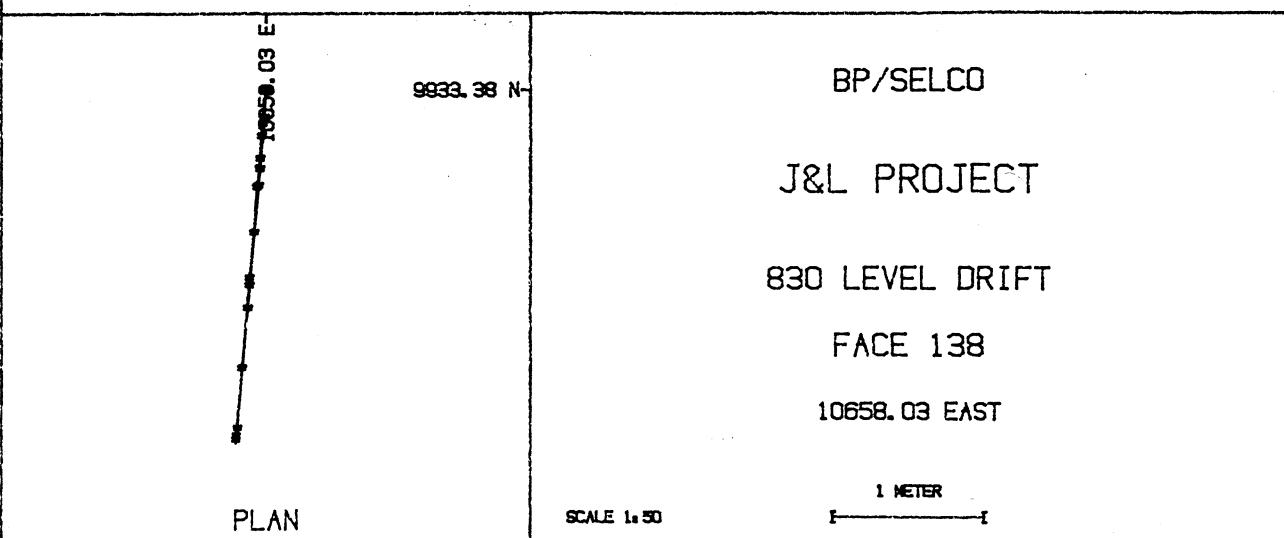
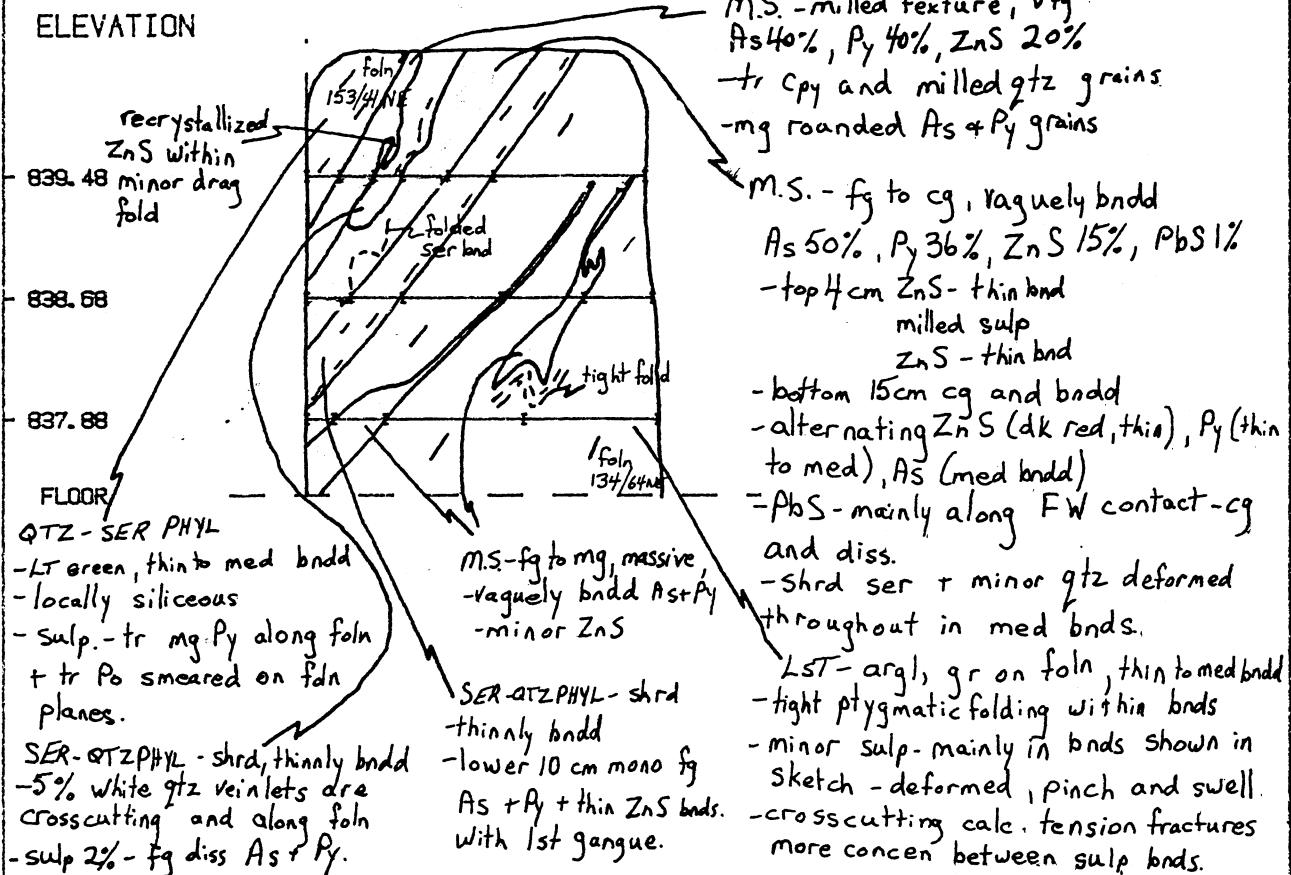
FACE# 138

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43515	.22	.24	1.83	1.020	7.7	1.2
43516	.23	4.68	15.30	4.340	169.0	9.2
43517	.18	.30	1.76	4.710	33.7	3.3
43518	.31	8.90	9.60	16.300	267.1	35.4
43519	.30	1.58	1.11	5.020	19.2	2.1
43520	1.00	.18	.25	3.180	7.7	1.9
43521	.30	8.59	3.72	14.800	307.8	30.2
43522	.34	.54	.68	2.980	18.1	3.1
43523	.65	.10	.05	.381	5.2	.5
43524	.55	6.47	8.69	2.390	139.5	3.1
43525	.45	.12	.13	.040	5.7	.3

SAMPLE NUMBERS						GEOLOGY CODES					
043515 043516 043517 043518 043519 043520						07 H 12 0 04 12 0 04 02 0					
043521 043522 043523 043524 043525						12 0 04 0 02 0 02 F					
043526 043527 043528 043529						04 0 12 0 02 0 02 F					

GEOLOGY

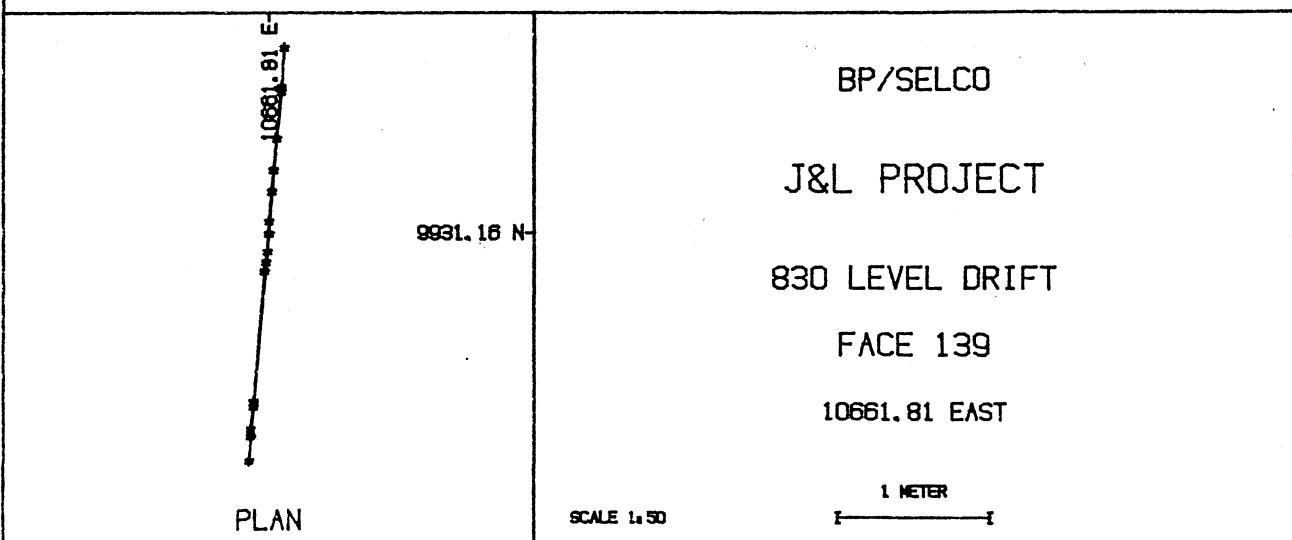
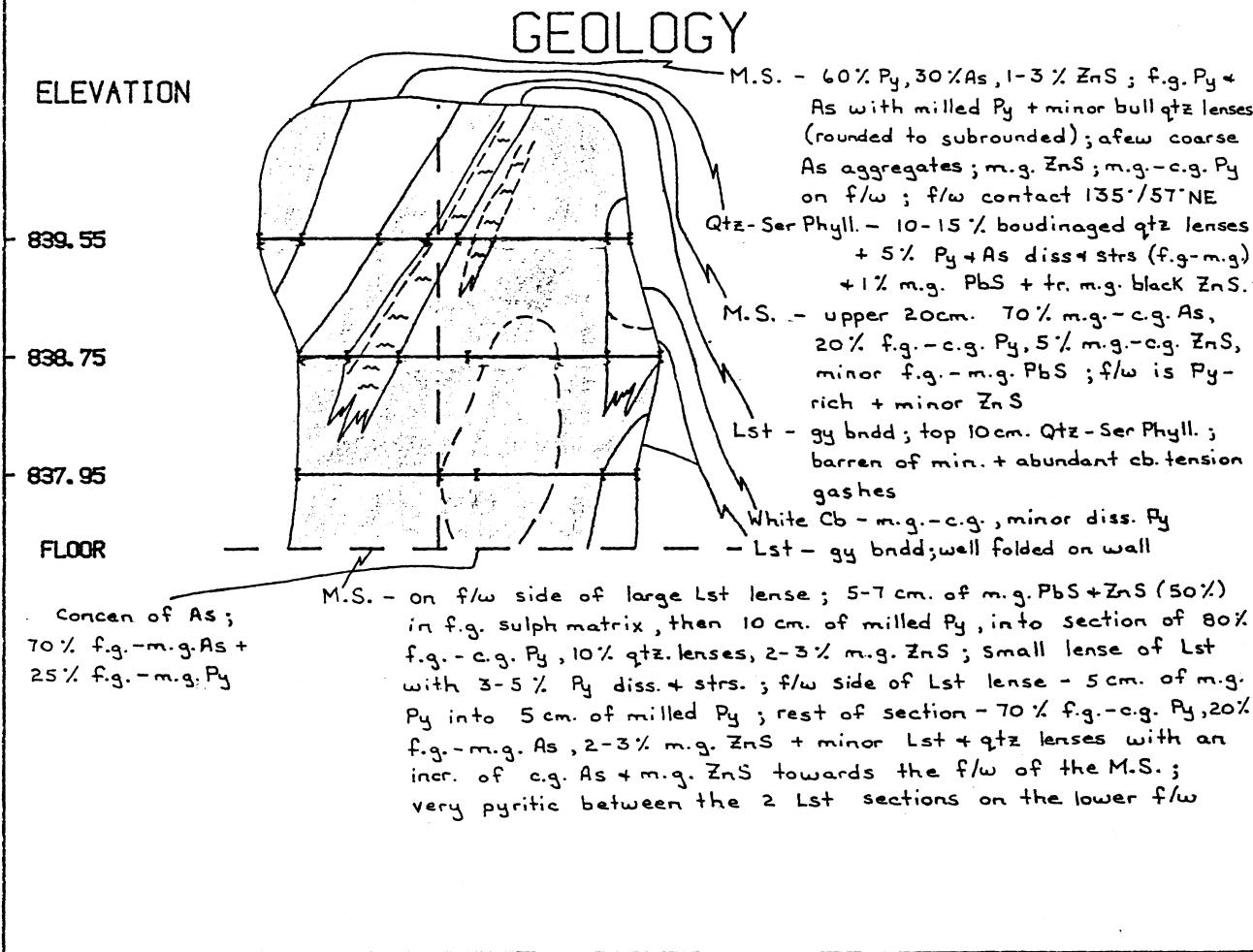
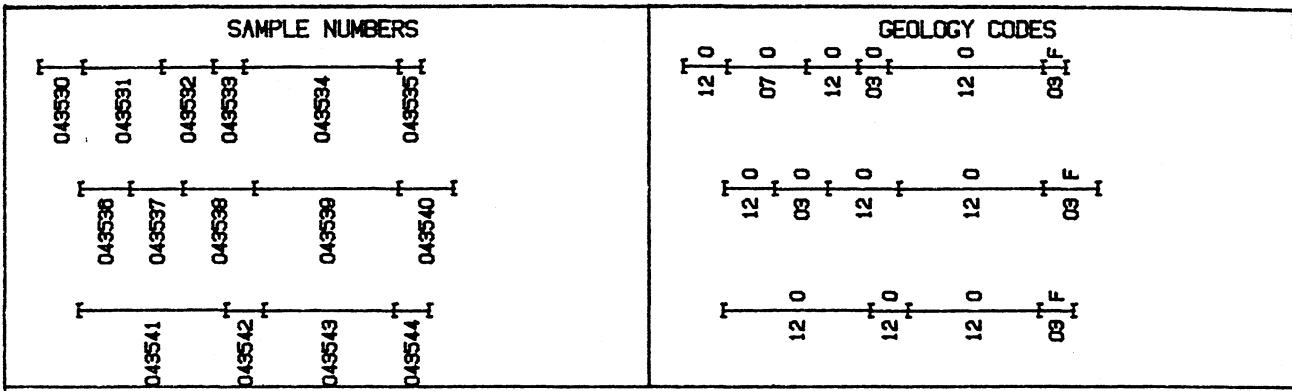
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 139

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43530	.29	6.09	17.60	5.010	276.9	9.7
43531	.52	.31	.38	3.030	15.9	1.9
43532	.34	4.56	4.26	24.200	104.5	38.1
43533	.20	.64	2.03	2.490	28.0	4.2
43534	1.02	3.08	10.10	8.310	41.3	8.4
43535	.15	.07	.31	.236	13.4	.3
43536	.33	3.26	6.72	20.100	196.7	38.5
43537	.35	.08	.09	.994	7.3	.9
43538	.47	.23	.58	.998	14.8	2.3
43539	.95	1.34	5.47	10.200	76.0	7.7
43540	.36	.13	.37	.638	7.6	.6

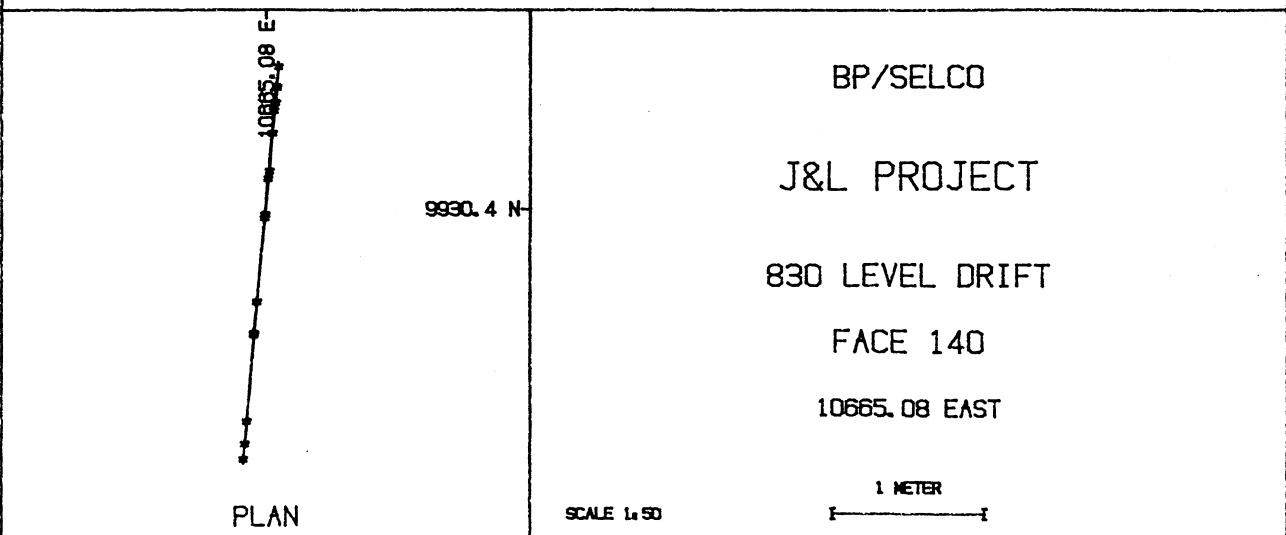
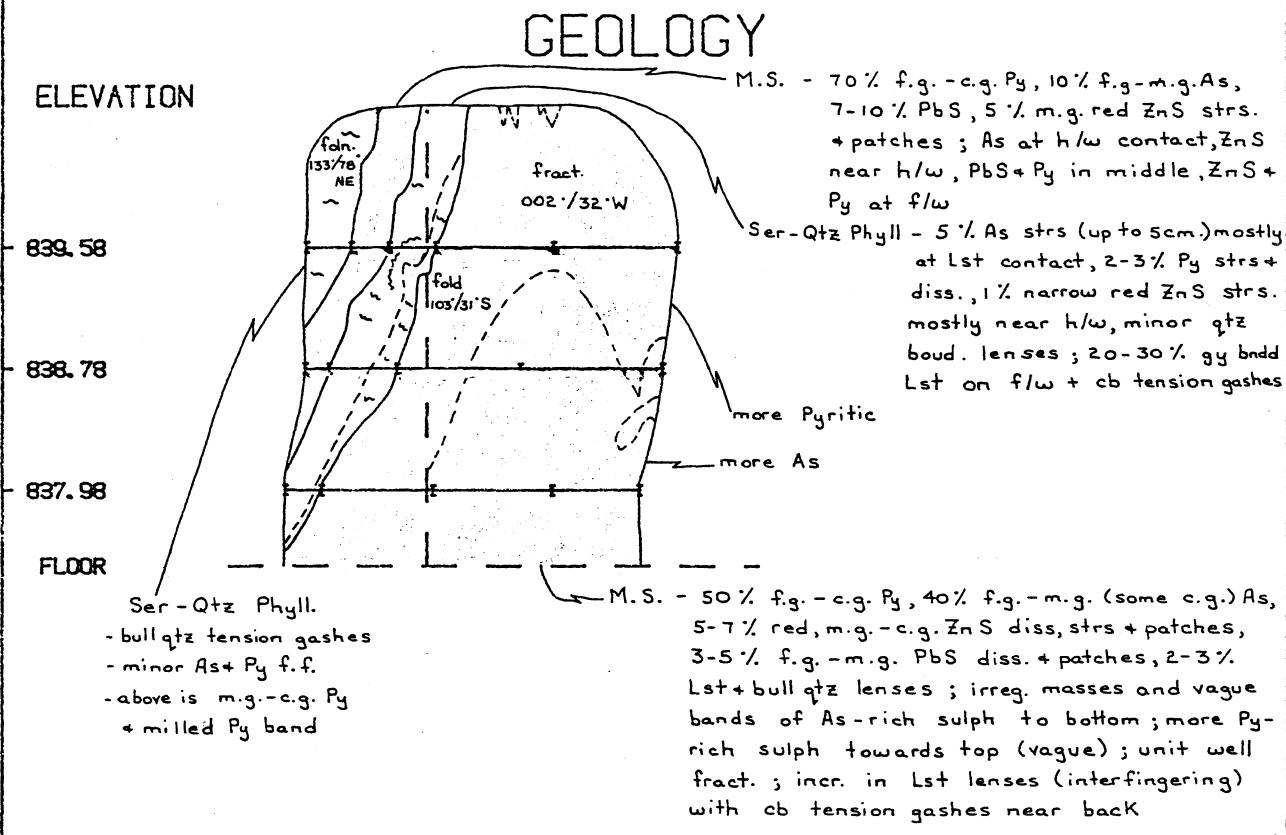
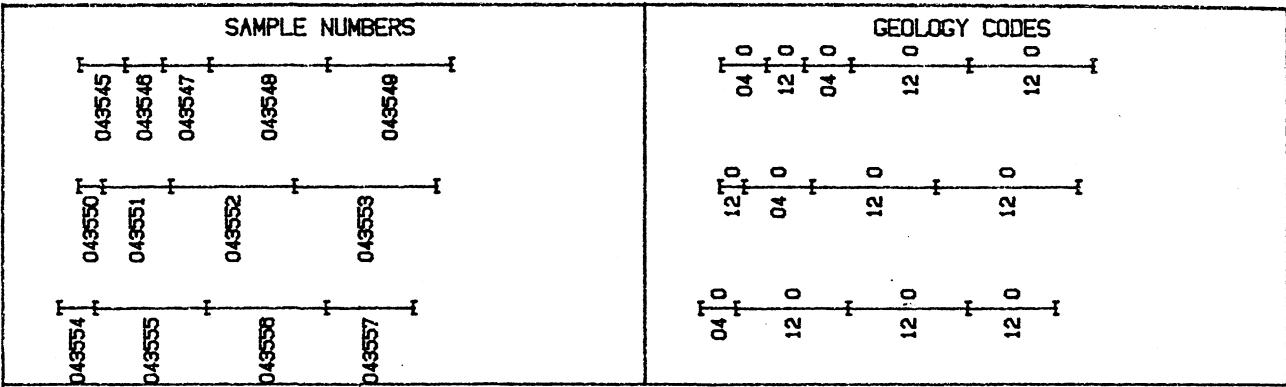


SEILCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 140

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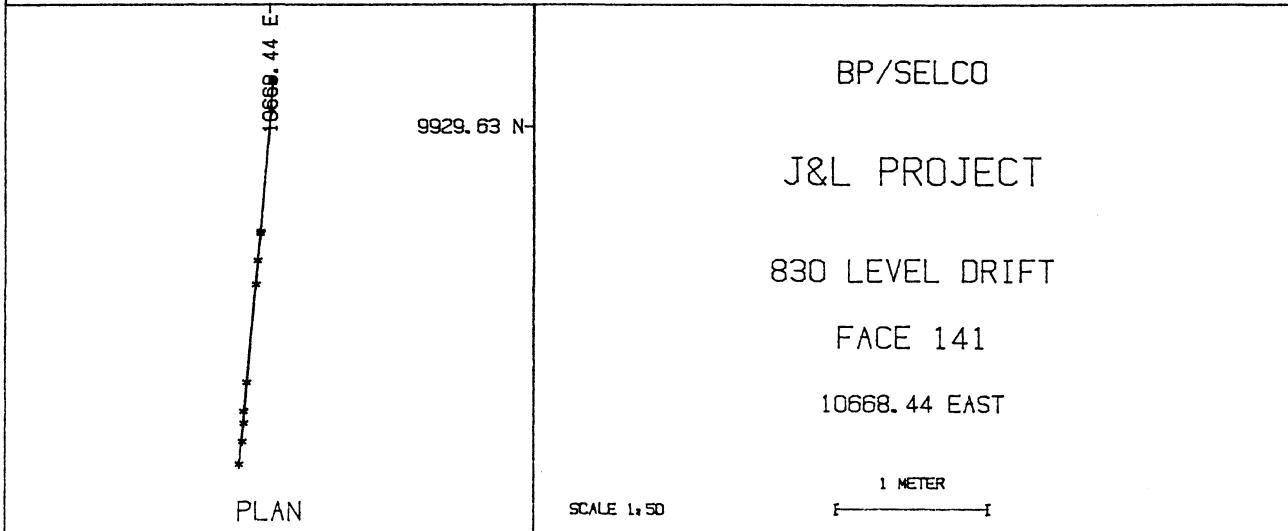
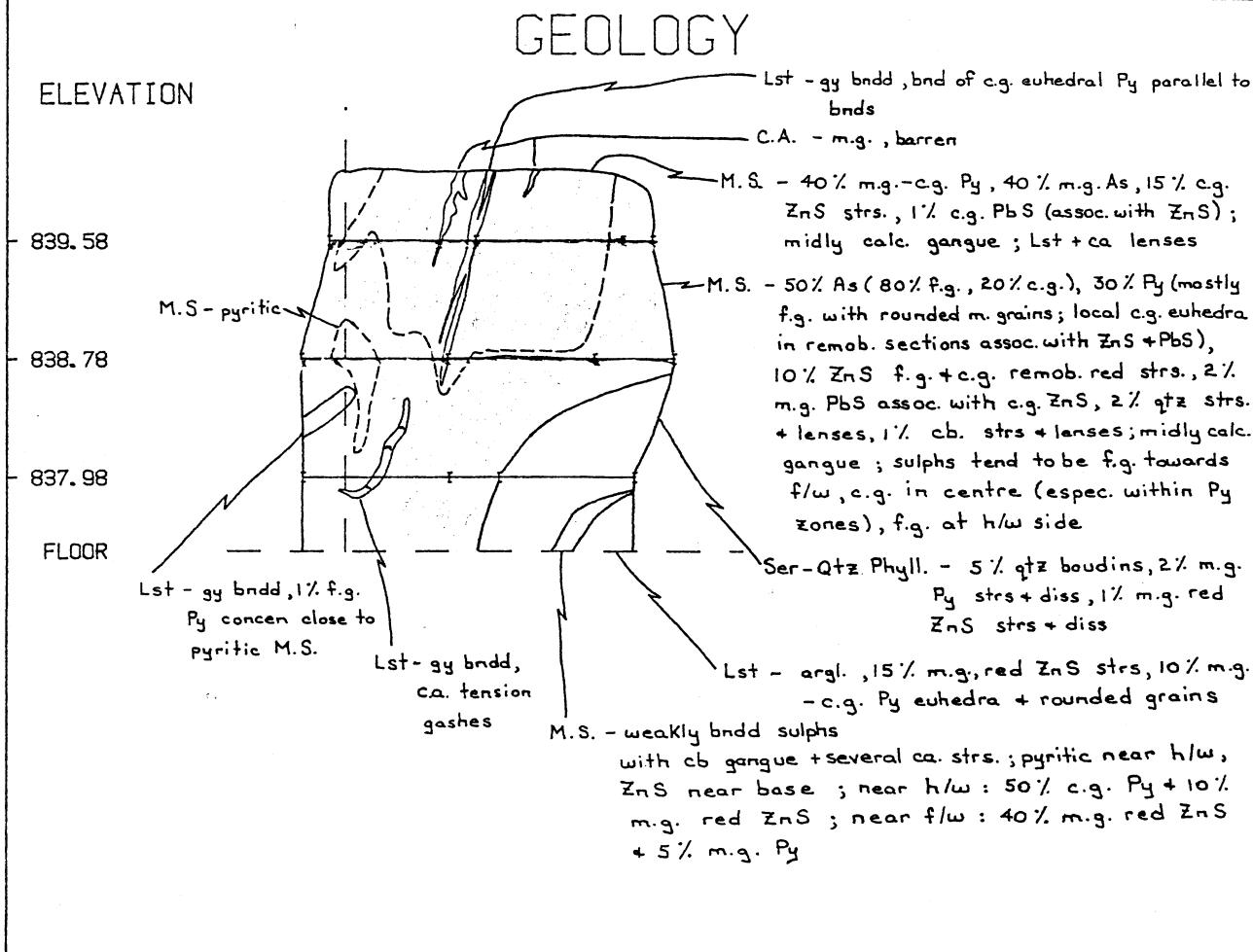
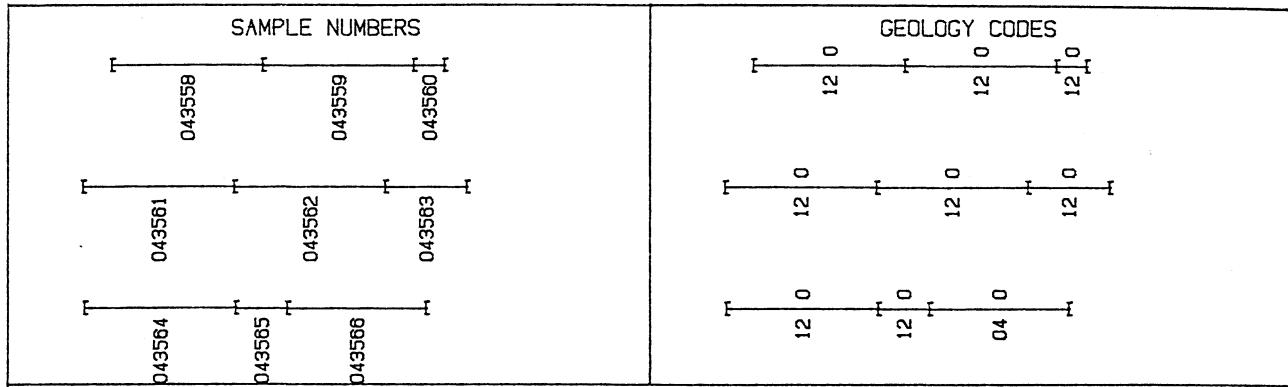
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43545	.30	.11	.18	.973	8.2	1.4
43546	.25	14.70	7.16	5.100	352.7	16.8
43547	.31	.70	.74	3.080	20.6	3.4
43548	.78	3.50	26.60	12.200	120.5	25.5
43549	.82	2.42	15.00	8.950	112.6	16.3
43550	.16	15.70	15.30	6.880	274.5	16.2
43551	.45	.41	.46	1.540	13.8	2.0
43552	.82	5.89	12.20	11.000	143.0	22.9
43553	.94	4.34	6.74	9.640	141.2	17.8



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 141

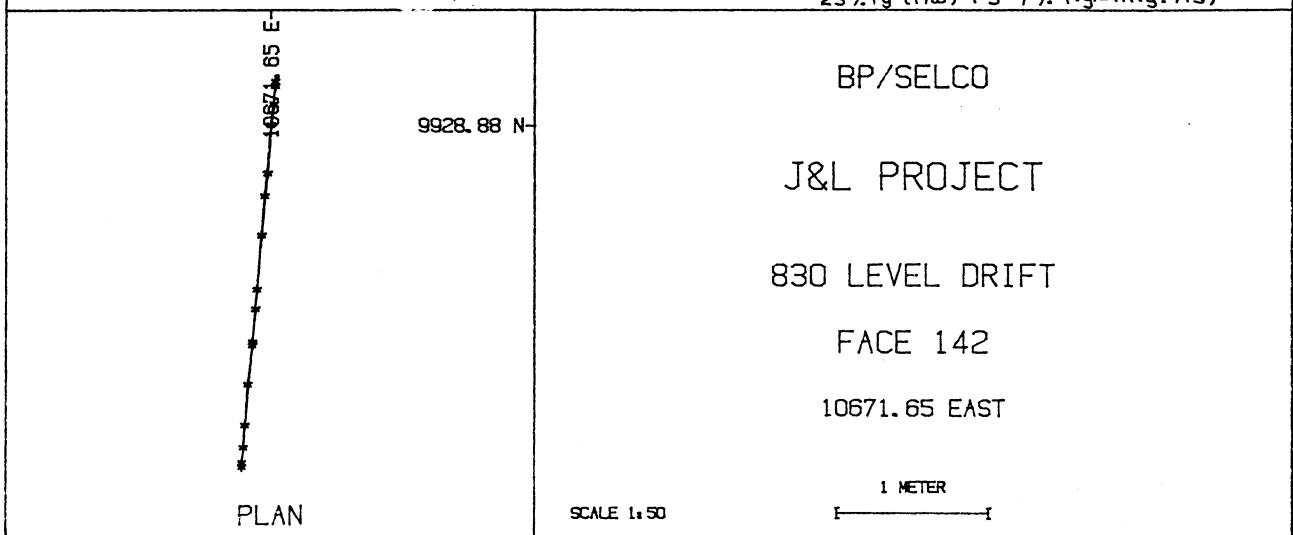
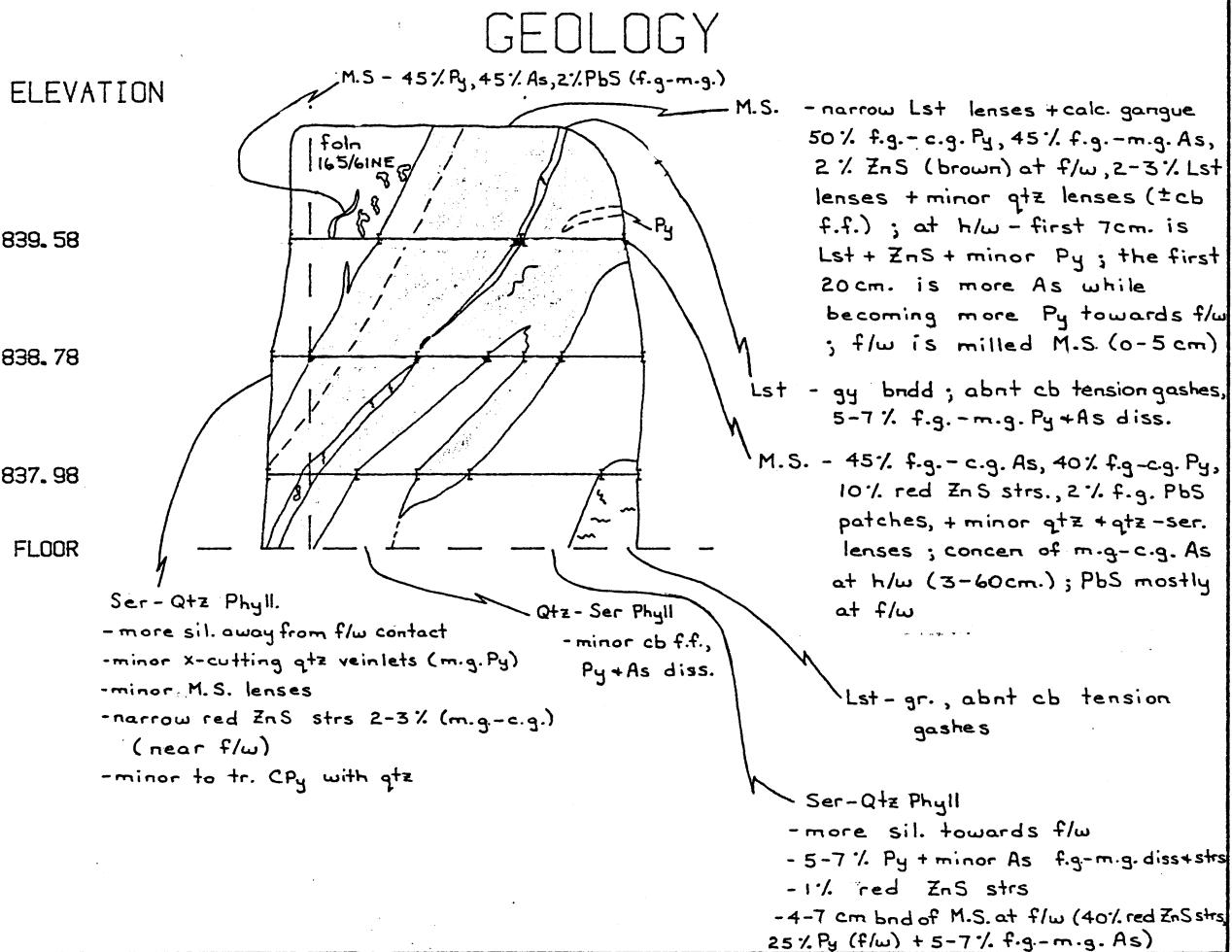
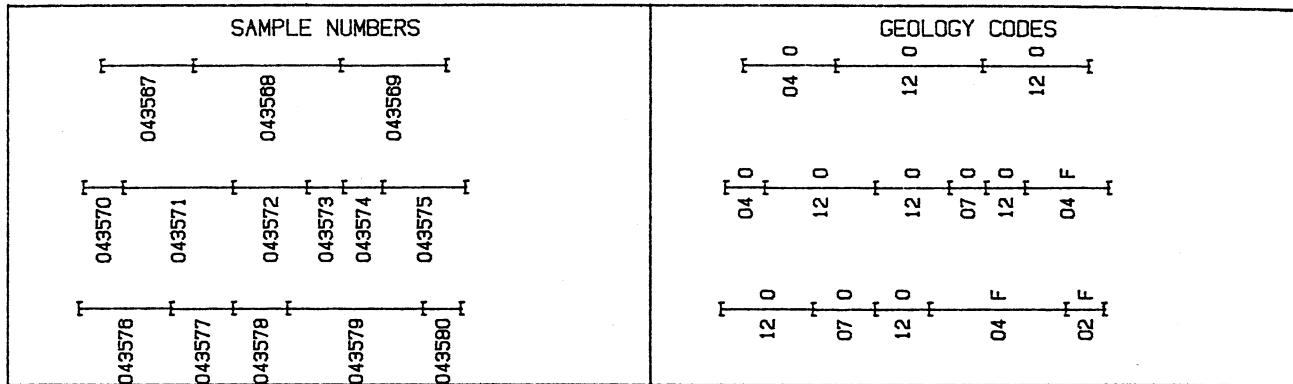
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43558	1.00	1.63	13.10	8.610	39.2	23.2
43559	1.00	2.78	9.20	8.270	56.8	19.3
43560	.20	.34	.80	15.900	8.6	22.9
43561	1.00	.98	4.78	8.340	30.9	8.2
43562	1.00	3.02	6.94	7.620	70.8	21.7
43563	.54	1.28	2.53	11.600	20.0	16.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 142

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43567	.61	1.14	2.13	1.520	17.2	4.0
43568	.77	.64	1.74	10.300	19.2	14.4
43569	.70	2.17	3.84	9.360	45.2	22.0
43570	.26	1.60	.48	3.300	23.4	2.6
43571	.73	.51	.36	8.490	15.8	14.4
43572	.49	1.08	2.70	15.600	29.7	21.7
43573	.24	.17	.07	4.050	5.5	2.0
43574	.26	8.55	3.72	9.490	208.7	13.4
43575	.55	.36	.31	1.390	16.7	.7



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 143

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43582	.24	.10	.08	2.290	1.4	.7
43583	1.00	.42	.88	6.050	14.4	7.5
43584	.64	2.84	1.92	8.590	73.7	12.7
43585	.43	.57	1.74	21.900	19.5	18.2
43586	.35	1.39	1.30	7.130	34.2	7.6
43587	.82	.41	.60	22.200	14.5	23.0
43588	.90	.16	1.99	21.400	8.3	17.1
43589	.25	1.37	4.26	13.600	37.0	12.4

SAMPLE NUMBERS				GEOLOGY CODES			
043582	043583	043584	043585	03 0	12 0	03 0	12 0
043586	043587	043588	043589	12 0	12 0	12 0	12 0
043590	043591	043592	043593	03 0	12 0	12 0	07 0

GEOLOGY

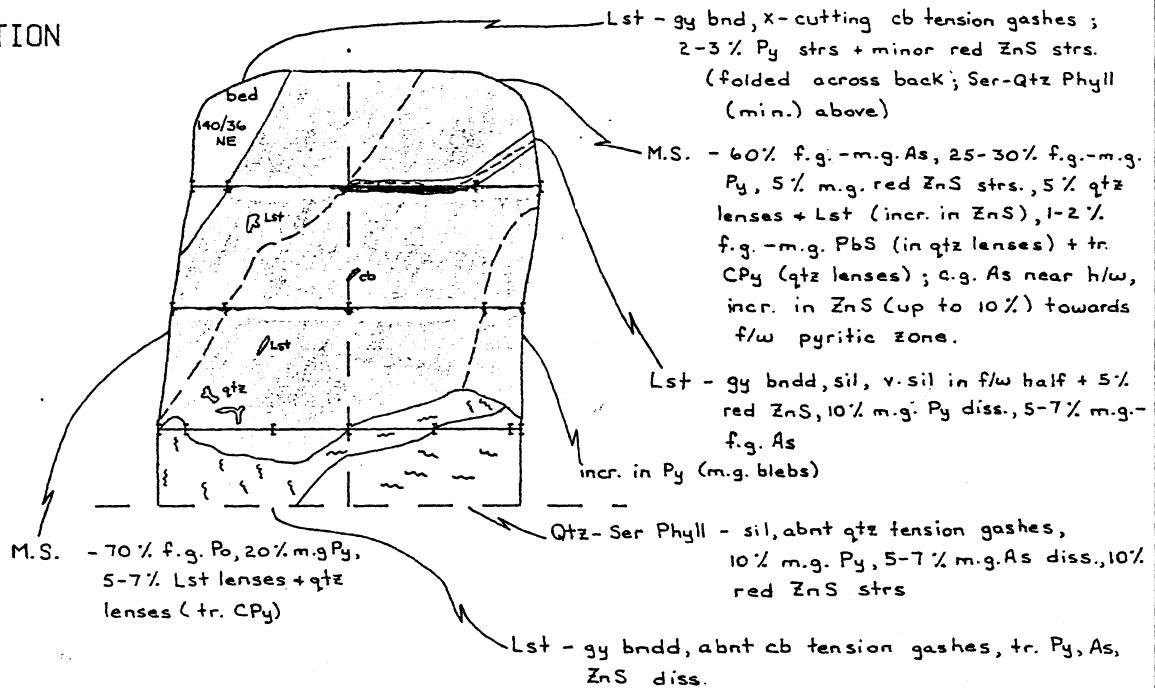
ELEVATION

839.63

838.83

838.03

FLOOR



10674.72 E
9926.57 N

BP/SELCO

J&L PROJECT

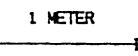
830 LEVEL DRIFT

FACE 143

10674.72 EAST

PLAN

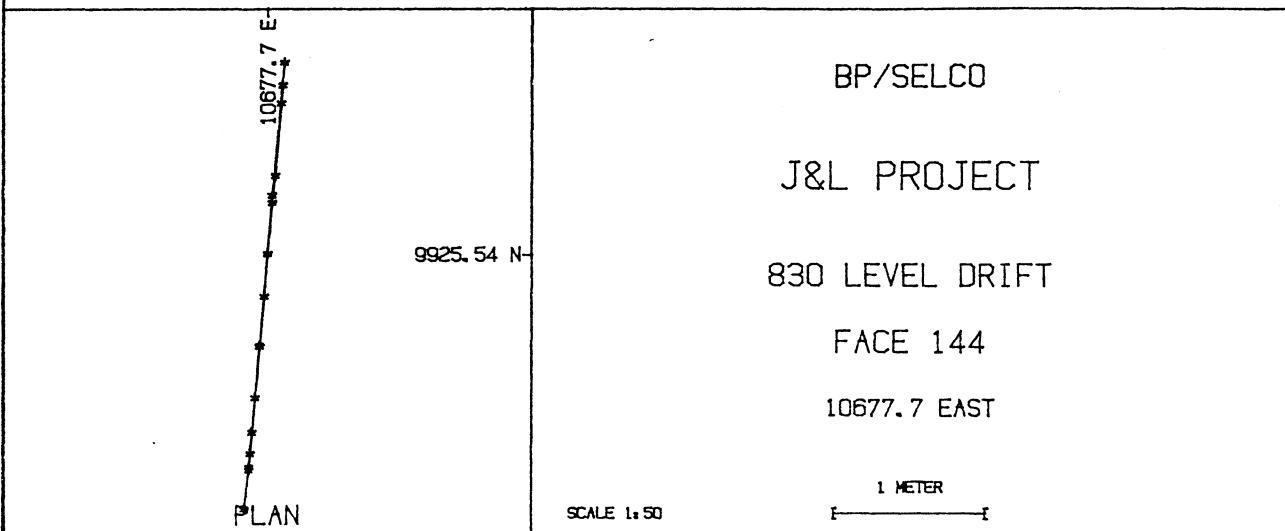
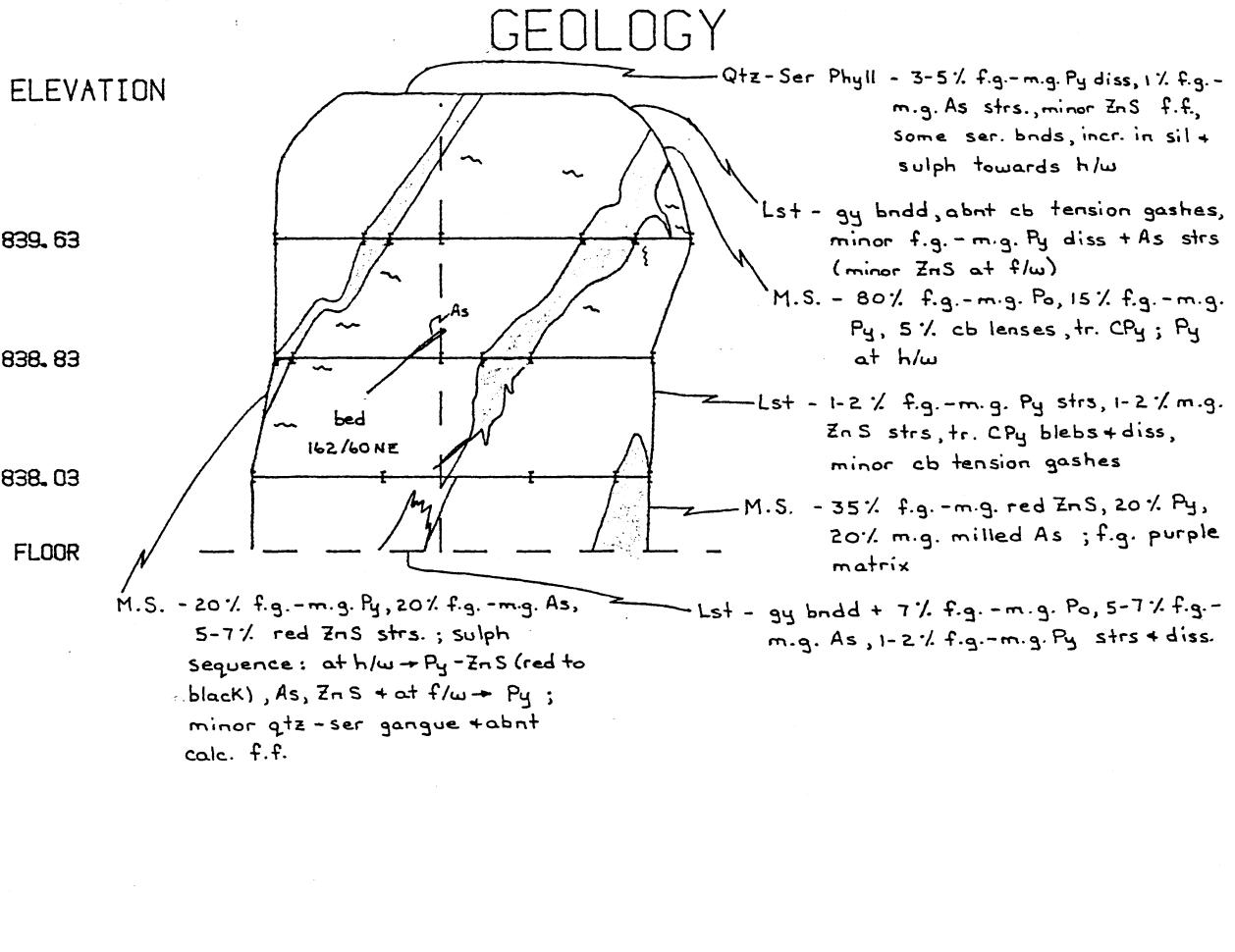
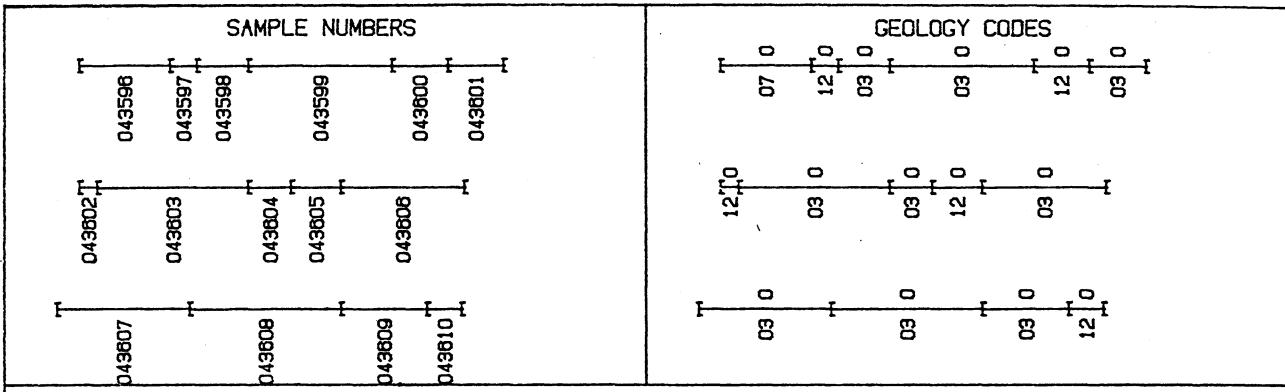
SCALE 1:50



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 144

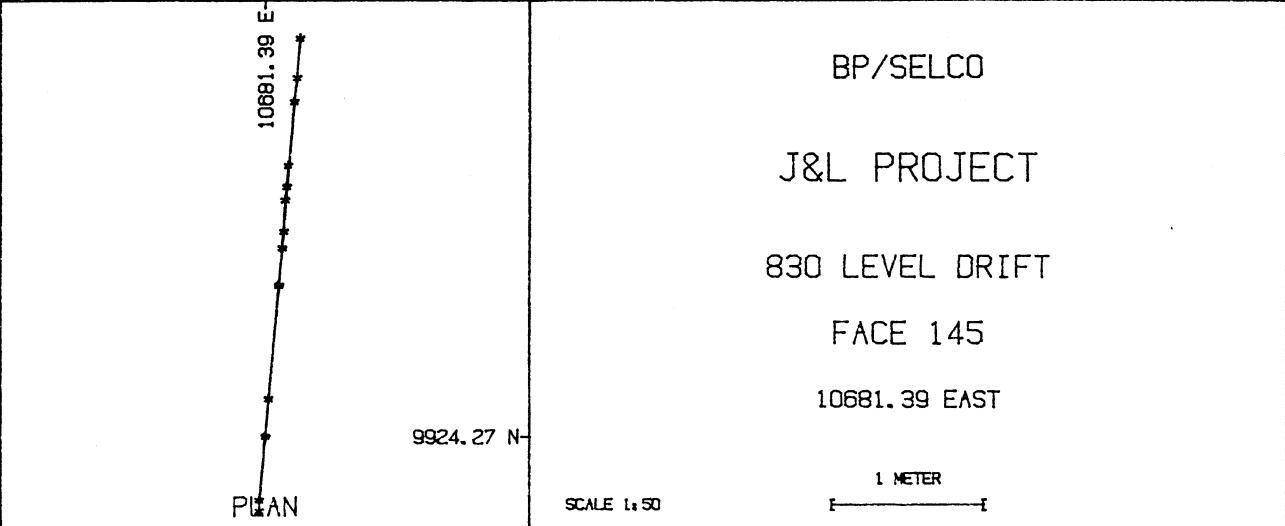
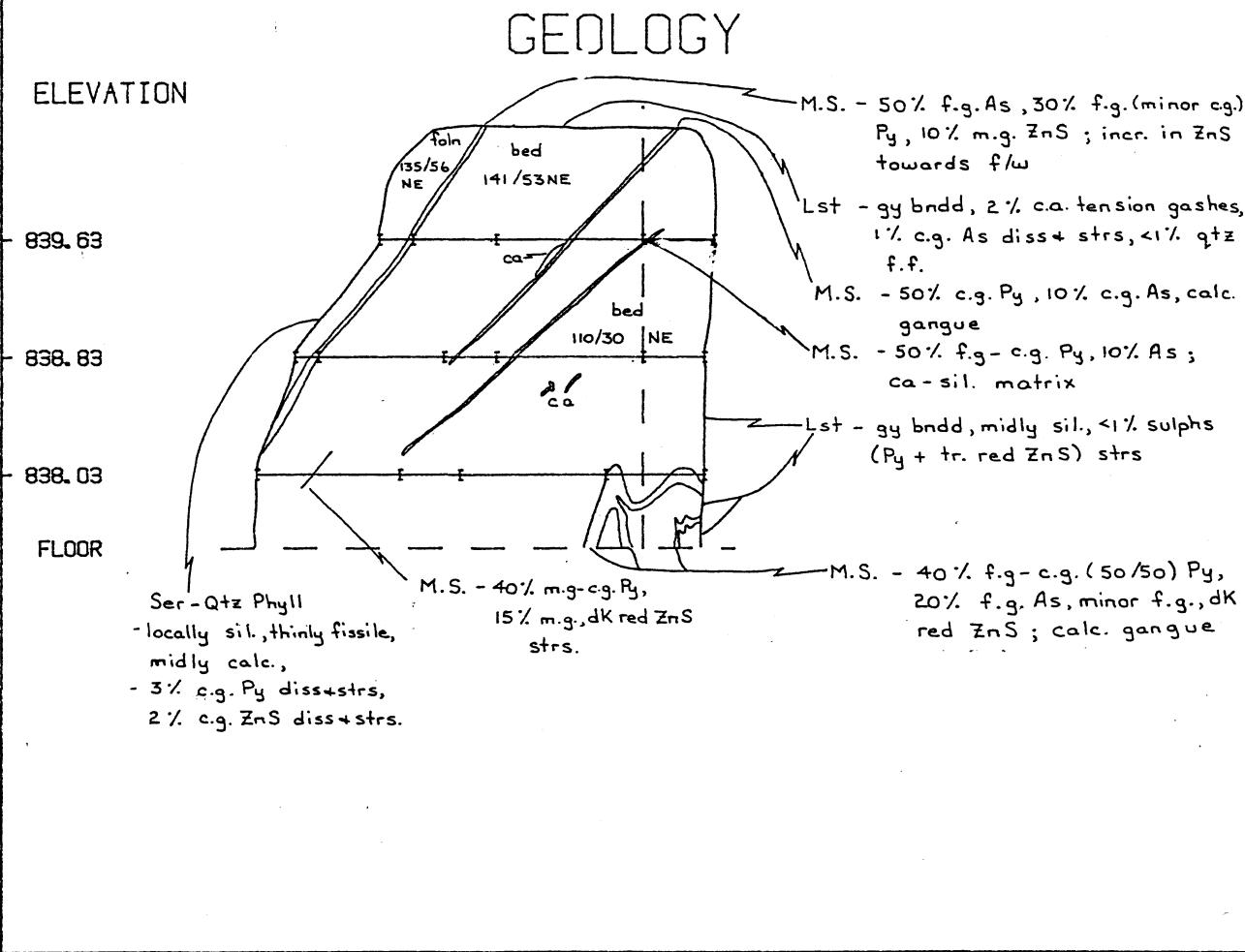
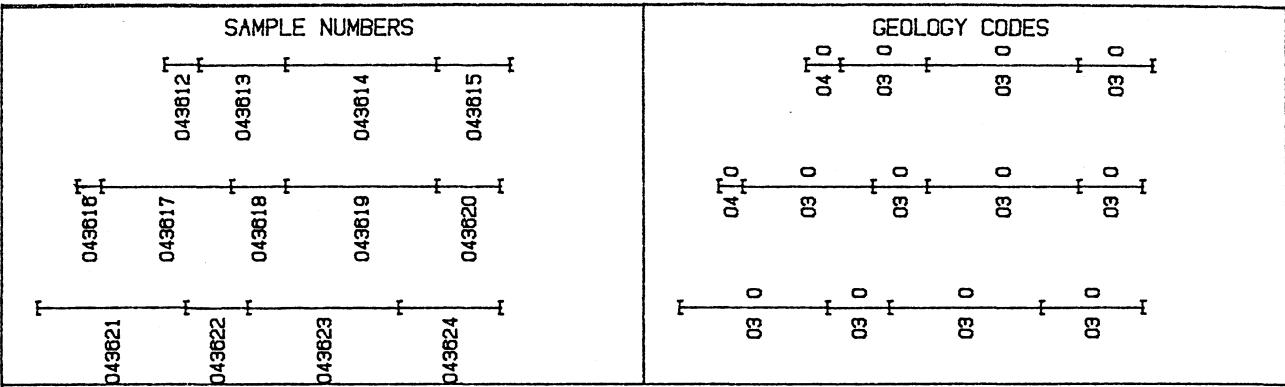
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43596	.60	.09	.54	.939	2.4	1.0
43597	.18	.79	1.30	13.100	18.1	14.8
43598	.34	.10	.10	.523	1.8	.9
43599	.95	.06	.04	.272	.6	.1
43600	.37	.11	.06	1.540	82.9	1.4
43601	.37	1.63	2.39	3.830	62.0	7.2
43602	.12	.65	4.26	9.680	29.3	15.9
43603	1.00	.13	.28	.322	4.3	.5
43604	.28	.06	.02	1.010	.9	.5
43605	.32	.07	.03	6.840	6.8	8.3
43606	.82	.28	.44	.676	4.1	.7



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 145

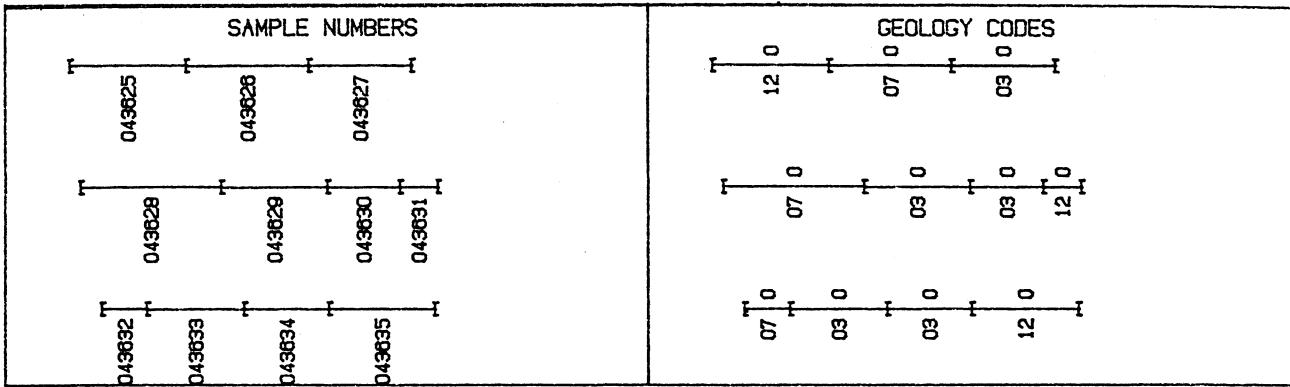
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43612	.23	.06	.01	2.220	3.2	.6
43613	.57	.01	.01	.125	1.1	.3
43614	1.00	.73	.39	2.620	12.4	3.4
43615	.49	.19	.01	12.900	12.6	8.6
43616	.16	.29	.07	1.120	.3	1.1
43617	.86	.16	.13	1.840	3.9	2.9
43618	.36	.16	.68	1.680	6.1	2.1
43619	1.00	.34	.44	1.000	5.8	1.7
43620	.42	.06	.13	.061	1.7	.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

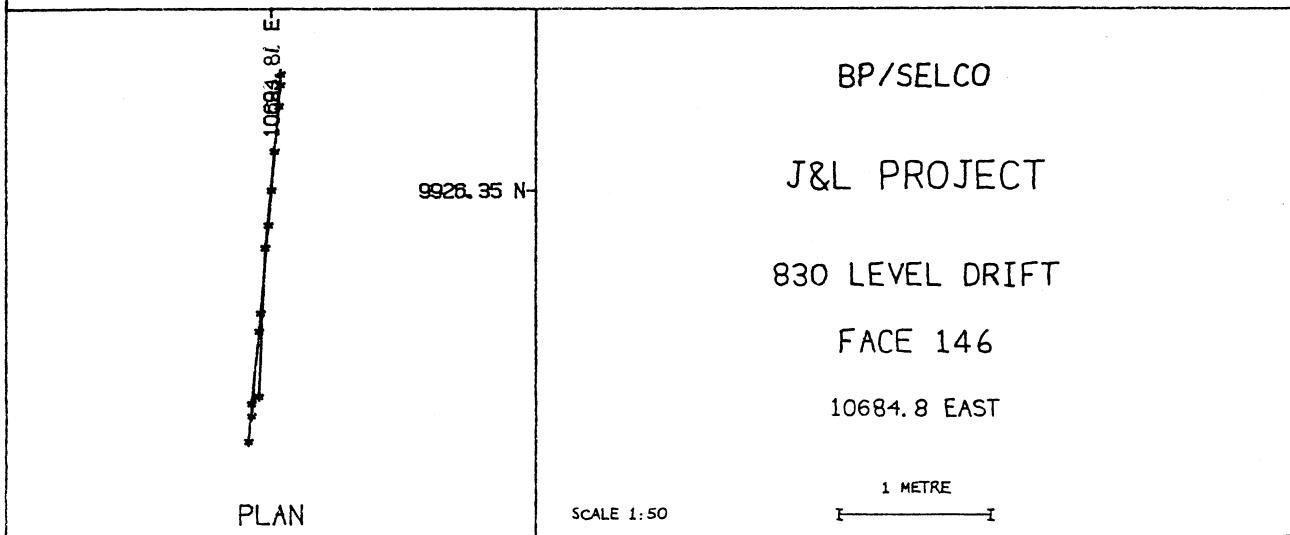
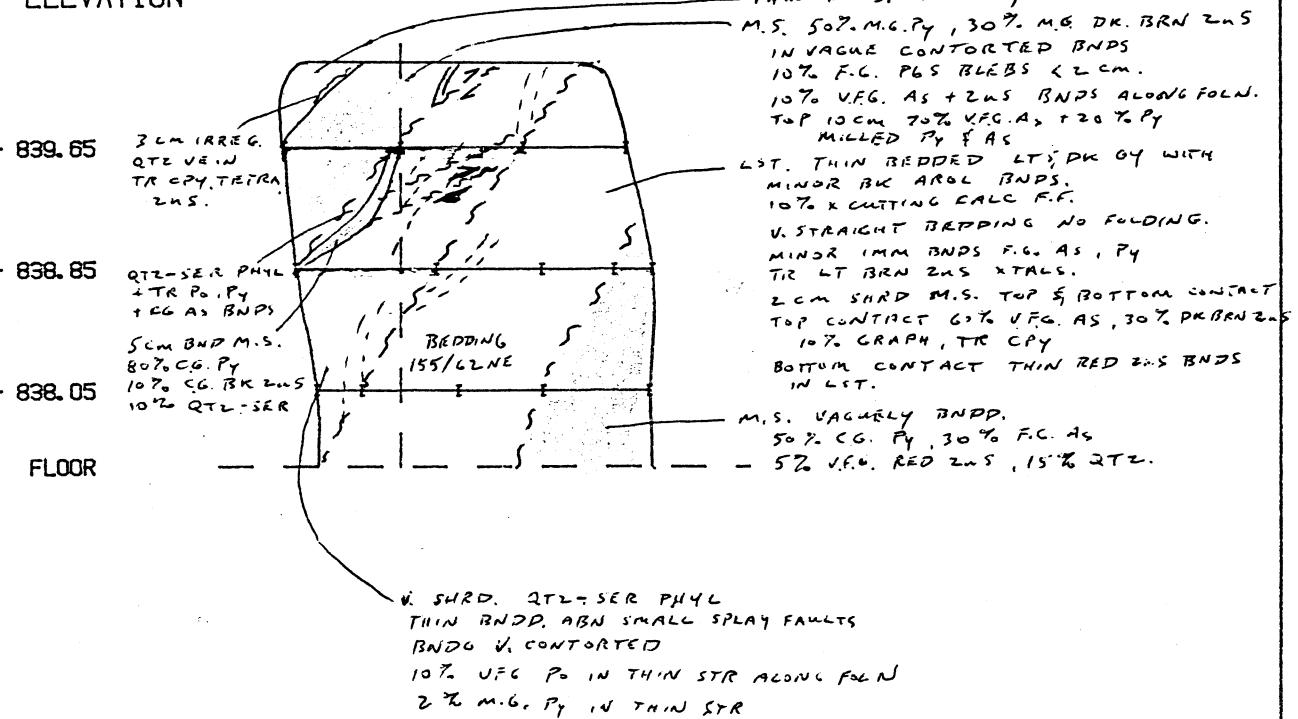
FACE# 146

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43625	.77	6.62	7.94	6.940	217.0	20.2
43626	.81	.23	.52	2.520	11.0	6.8
43627	.68	.48	.19	1.090	.8	1.2
43628	.93	.90	3.65	11.600	44.6	26.7
43629	.70	.03	.06	.315	.3	.5
43630	.48	.10	.11	.579	1.3	.7
43631	.25	1.51	.88	2.620	29.8	3.1



GEOLOGY

ELEVATION

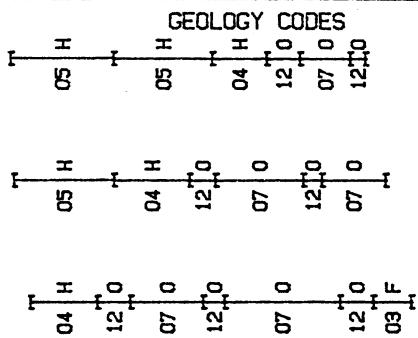
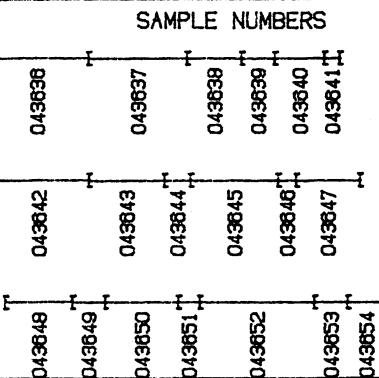


SELCO DIV. -- BP RESOURCES

J & L PROJECT. B.C.

FACE# 147

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43636	.68	.01	.02	.139	4.5	.3
43637	.65	.01	.01	.052	3.8	.2
43638	.36	.01	.02	.011	4.8	.1
43639	.22	8.07	12.00	4.720	216.0	9.0
43640	.33	.10	.11	1.270	4.8	.1
43641	.10	.36	3.15	11.000	33.9	9.8
43642	.66	.01	.04	.143	5.1	.3
43643	.50	.05	.06	.101	5.8	.3
43644	.17	8.22	13.50	5.410	251.3	12.1
43645	.58	.10	.16	2.320	9.6	.7
43646	.12	.46	2.53	13.500	36.7	8.4
43647	.42	.04	.07	.189	4.5	.5



GEOLOGY

ELEVATION

839.65

CHL-SER; QTL PHYL
DK GRN. V. SHRD.
ABN. SMALL SPLAY FAULTS MINOR GOUGE
20-40% VCG. QTL. LENSES W 3MM
POL. XTHLS, TR PO & PY F.G.
LOCALLY V.CG. MUSC IN QTL.
TR FUSCHITE?

M.S. BNDD. 40% CG. PY, 20% MG. BRN ZWS
20% F.G. AS, 20% QTL-SER PHYL.

838.85

QTL-SER PHYL.
THIN BNDD, WEAKLY SHRD.
5-10% VFG. PO IN 1MM STR ALONG FOLN.

838.05

M.S. MILLED?
40% VFG. RED ZWS
40% VFG. AS
20% MG. PY
WEAKLY CALCAREOUS
LST. LT. DK CY RNDD, MINOR
- THIN BK ARG. BEDS.
2% CG. PY DISS.

FLOOR

QTL-SER PHYL.
THIN BNDD WEAKLY SHEARED
MINOR SEC QTL LENSES
10% CG. PY MONO 2MM BNDS
5% F.G. RED ZWS 1MM BNDS
10% CG. AS 2MM BNDS

M.S. MILLED - 60% VFG AS, 20% RED MG. ZWS
20% 1-2MM PY MILLED
TOP CONTACT 3CM MORE ABN RED ZWS
BOTTOM CONTACT MG BLEBS OF PY IN THICKER AREAS

SHRD SER-QTL PHYL, THIN BNDD.
MINOR X CUTTING 2MM QTL FF.
MINOR V. THIN BNDS F.G.. AS & PO

10688.1 E

9926.49 N

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 147

10688.1 EAST

PLAN

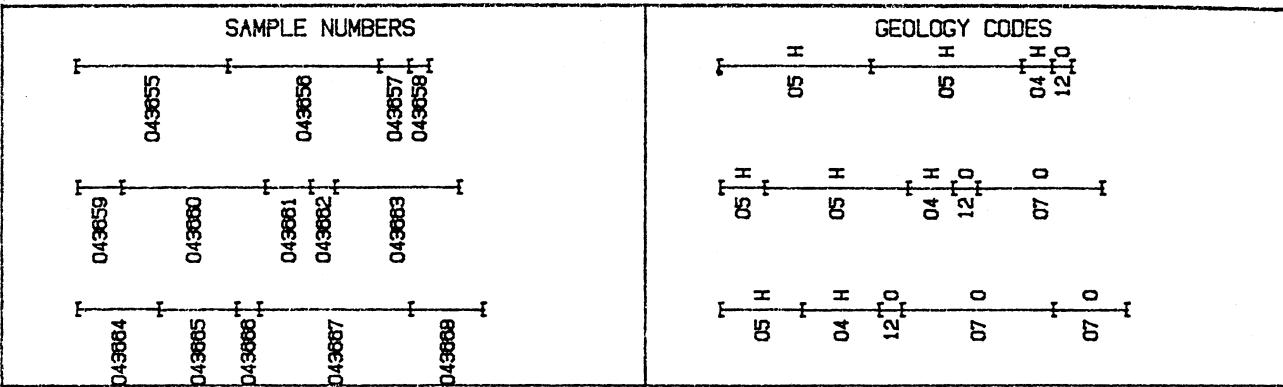
SCALE 1:50

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

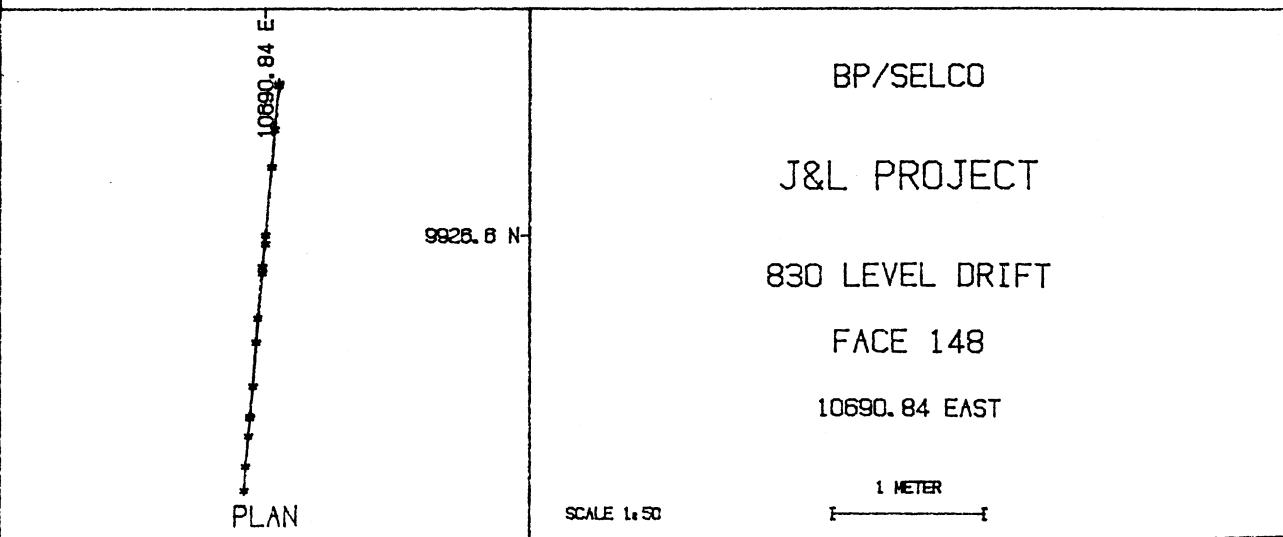
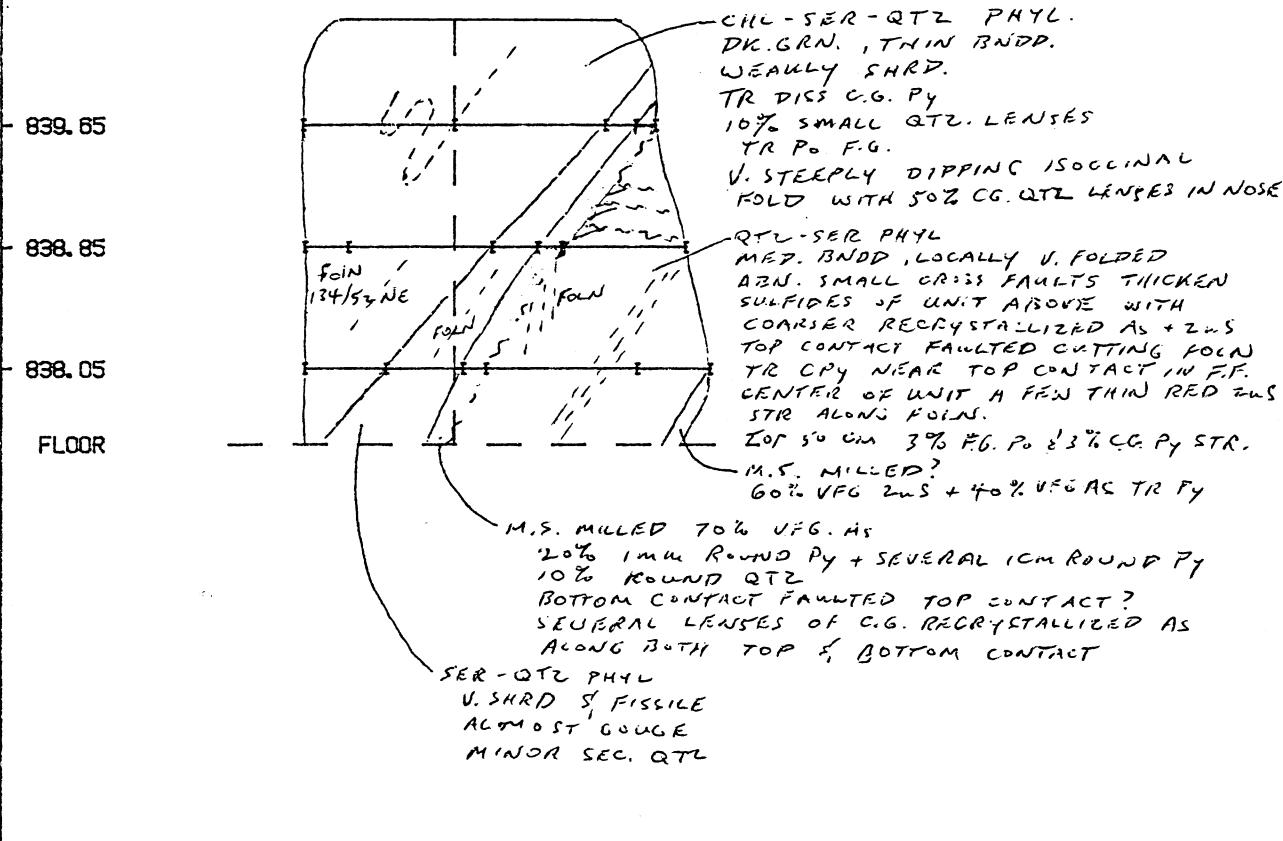
FACE# 148

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43655	1.00	.01	.03	.039	3.1	.2
43656	1.00	.01	.03	.069	6.2	.1
43657	.20	.02	.03	.062	4.5	.5
43658	.13	8.38	13.90	6.770	241.0	10.8
43659	.29	.06	.11	.034	5.5	.1
43660	.95	.01	.02	.024	4.1	.1
43661	.30	.01	.01	.060	3.1	.3
43662	.16	8.39	11.30	10.900	196.8	17.7
43663	.82	.23	.38	.198	11.0	.8



GEOLOGY

ELEVATION

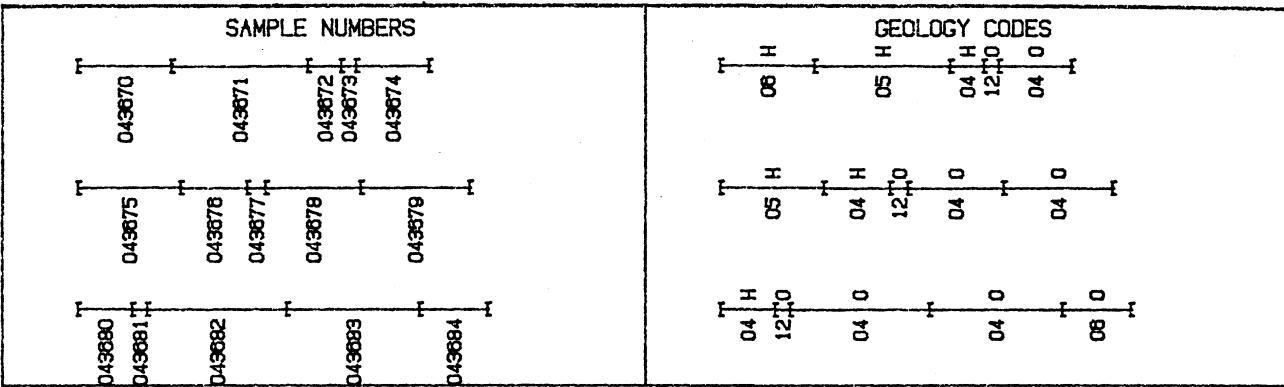


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 149

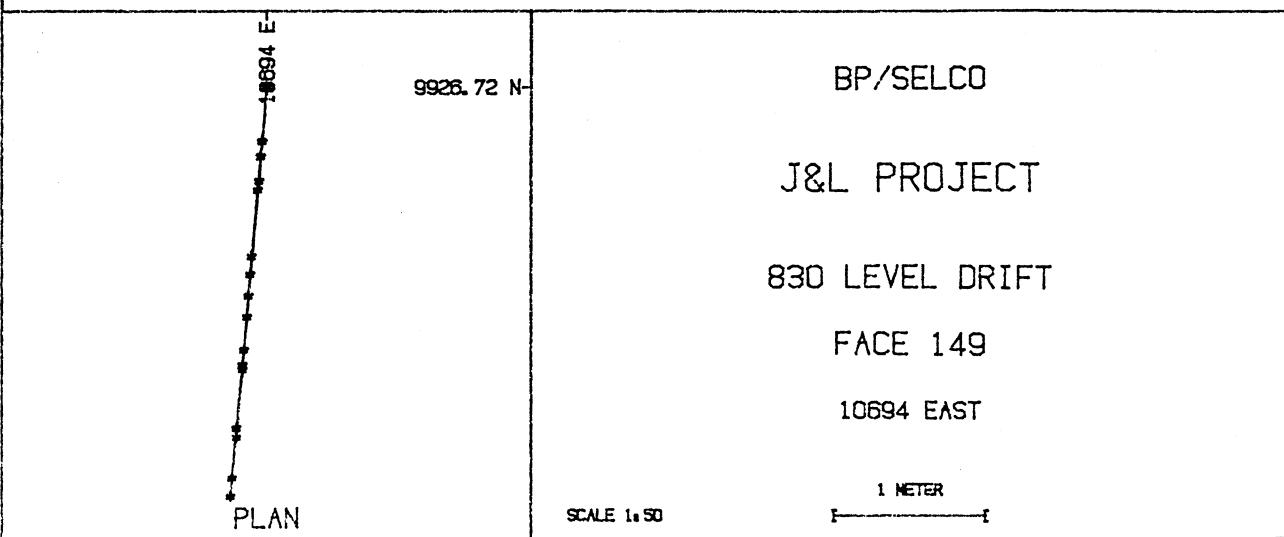
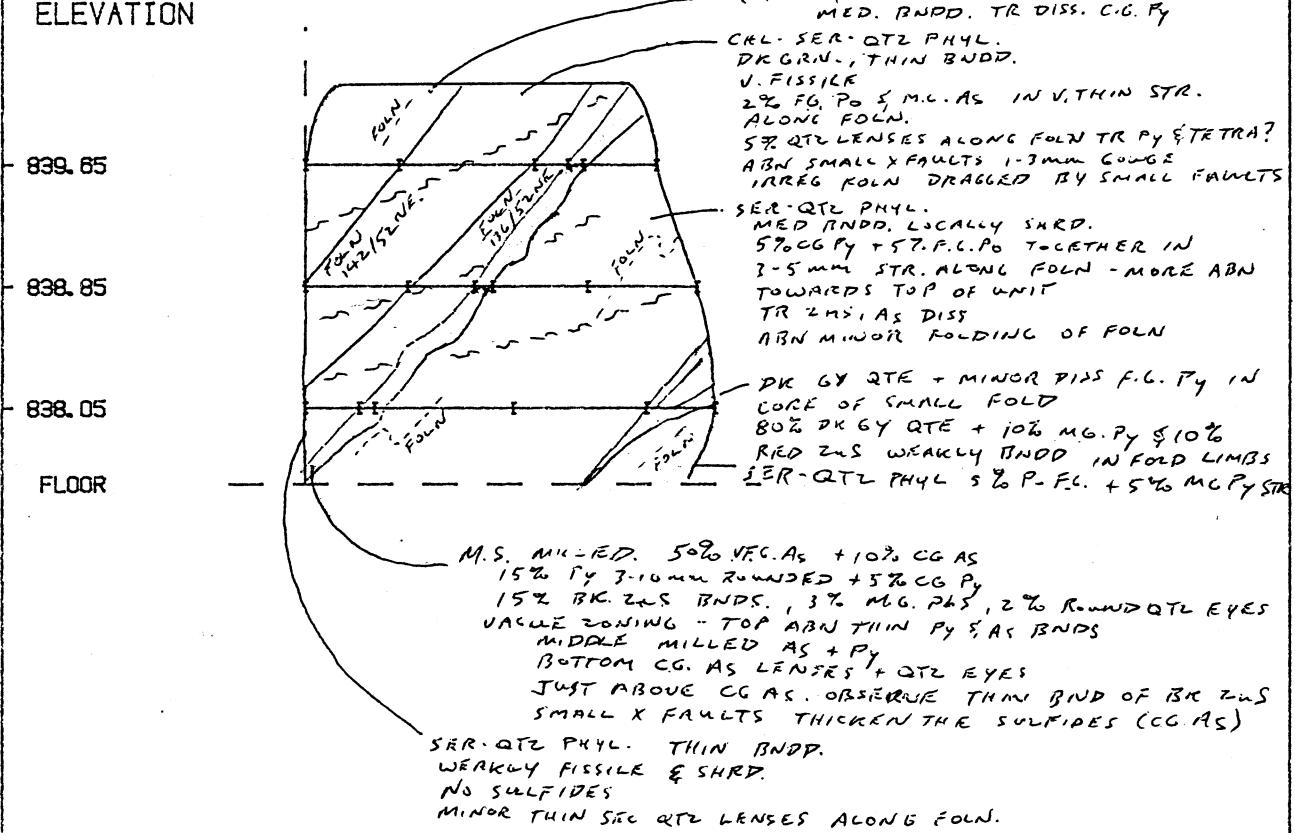
* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43670	.62	.01	.01	.014	.3	.1
43671	.90	.01	.01	.063	.3	.3
43672	.22	.01	.01	.004	.5	.2
43673	.10	7.09	10.40	20.000	202.0	16.0
43674	.48	.11	.15	.326	7.3	.2
43675	.68	.04	.04	.162	5.2	.3
43676	.44	.06	.05	.155	3.1	.3
43677	.12	8.97	14.50	5.340	270.4	12.7
43678	.63	.11	.15	.320	5.2	.3
43679	.72	.05	.10	.031	7.5	.1



GEOLOGY

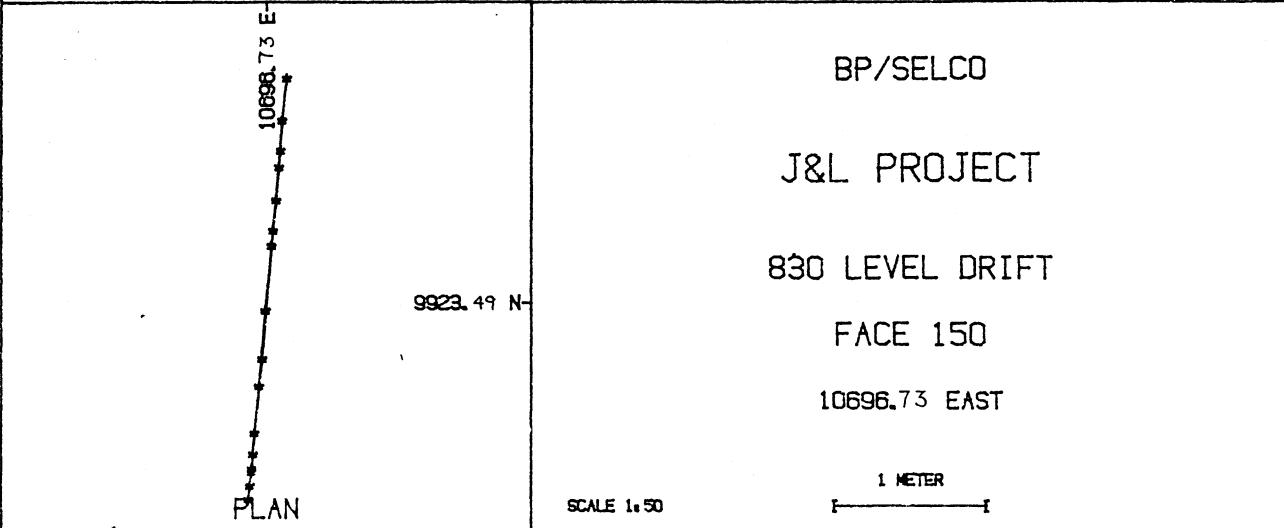
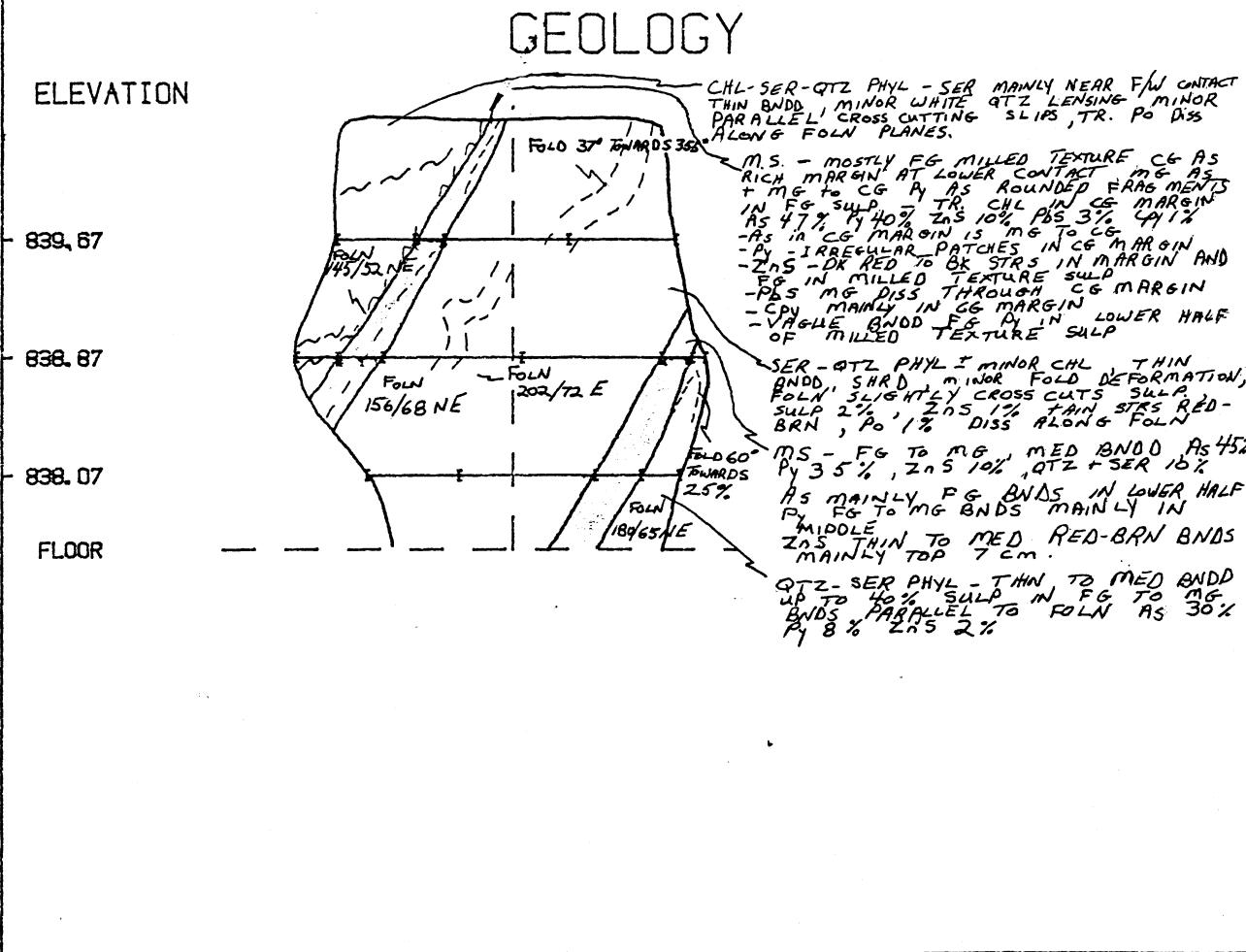
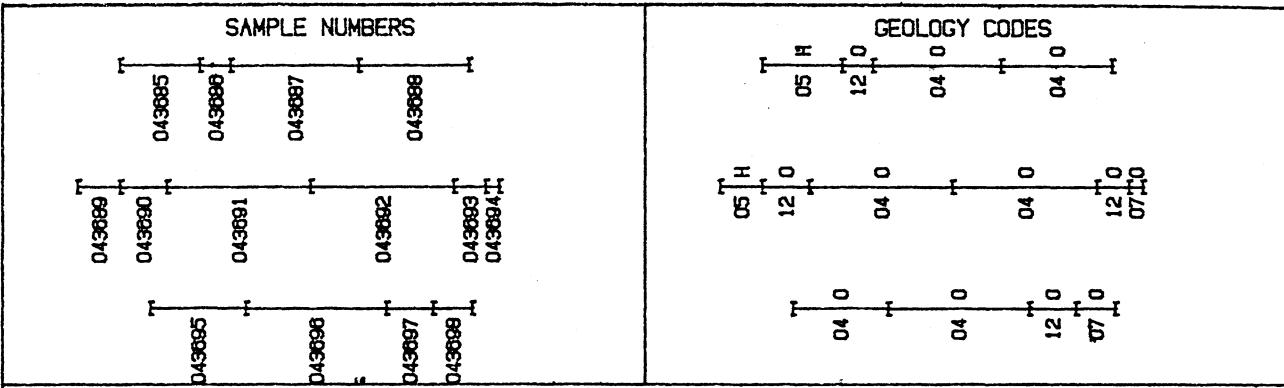
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 150

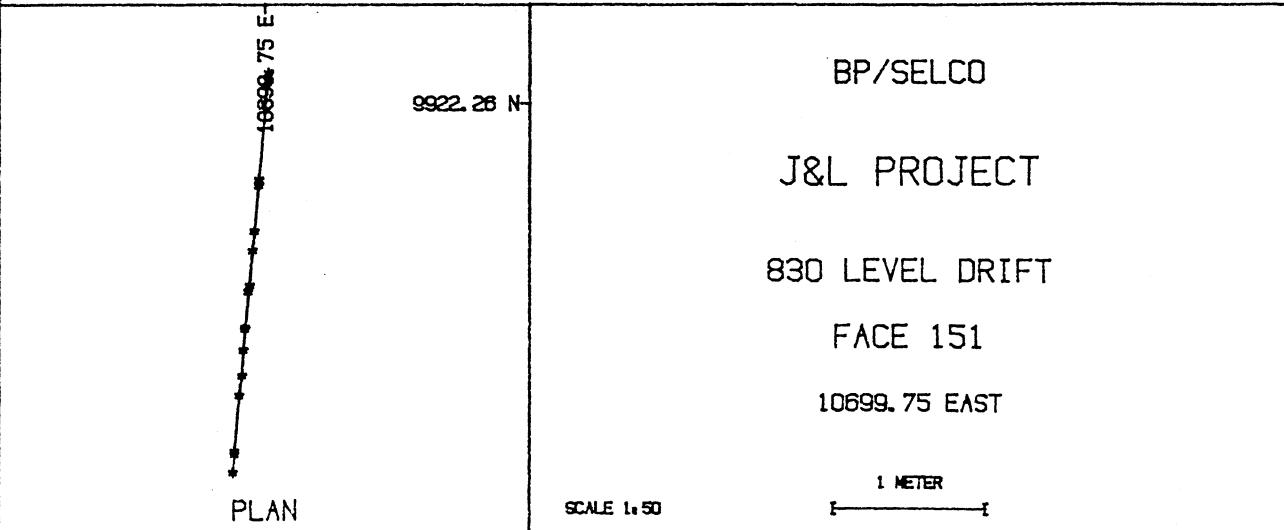
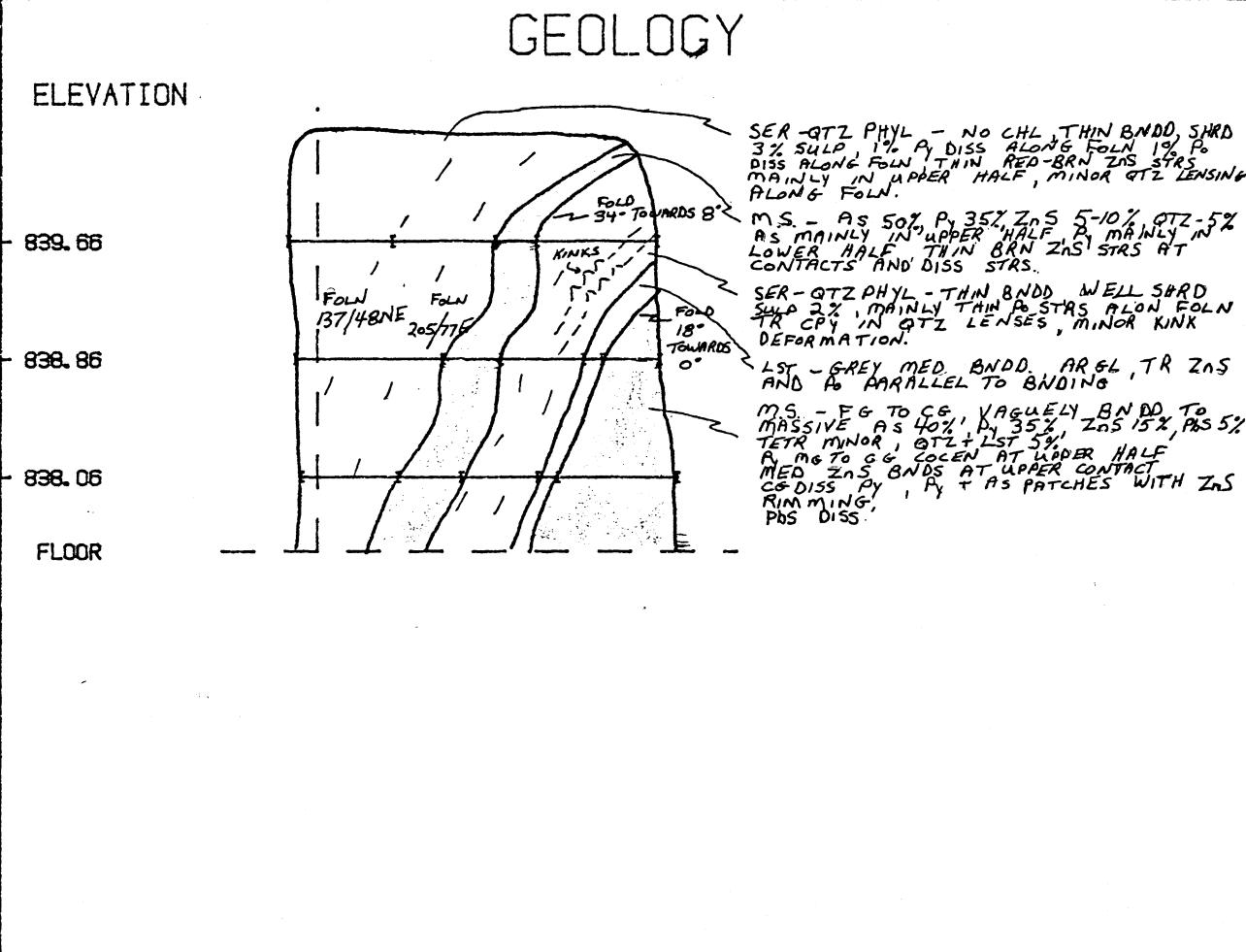
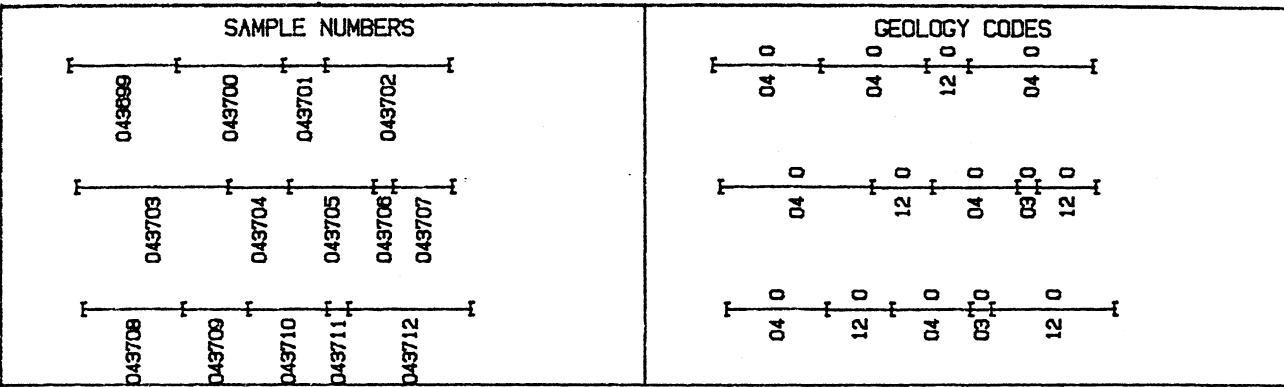
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43685	.53	.01	.02	.120	.3	.2
43686	.20	8.27	12.00	7.040	228.3	11.0
43687	.85	.91	1.70	.180	33.1	.5
43688	.73	.06	.06	.078	7.2	.3
43689	.28	.01	.02	.106	.3	.3
43690	.31	7.88	12.00	9.340	211.9	8.2
43691	.95	.10	.14	.157	3.8	.3
43692	.95	.01	.02	.033	.3	.2
43693	.21	.70	1.35	8.630	35.9	6.6
43694	.09	.58	.72	4.160	27.6	2.6



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 151

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43699	.71	.09	.06	.156	.5	.2
43700	.70	.01	.01	.378	.3	.1
43701	.28	1.92	2.80	5.580	92.0	8.1
43702	.82	.04	.04	.094	1.0	.4
43703	1.00	.01	.01	.032	.3	.1
43704	.40	2.93	3.50	17.400	103.2	22.3
43705	.56	.11	.08	.555	4.8	.7
43706	.13	.07	.05	.135	3.1	.3
43707	.39	4.12	7.40	6.970	68.2	10.6



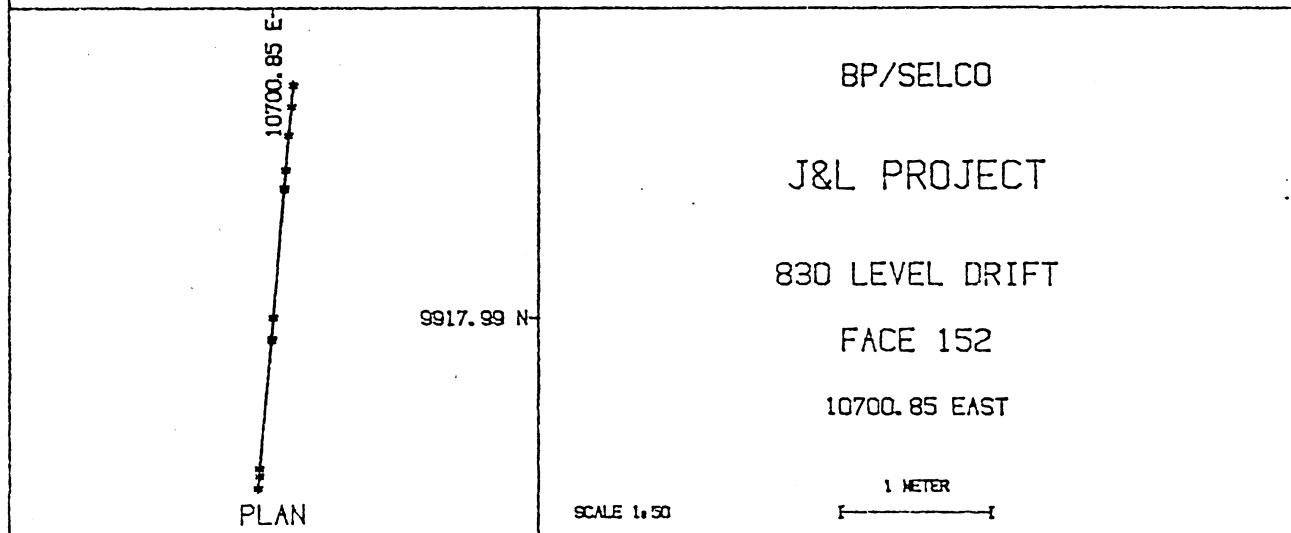
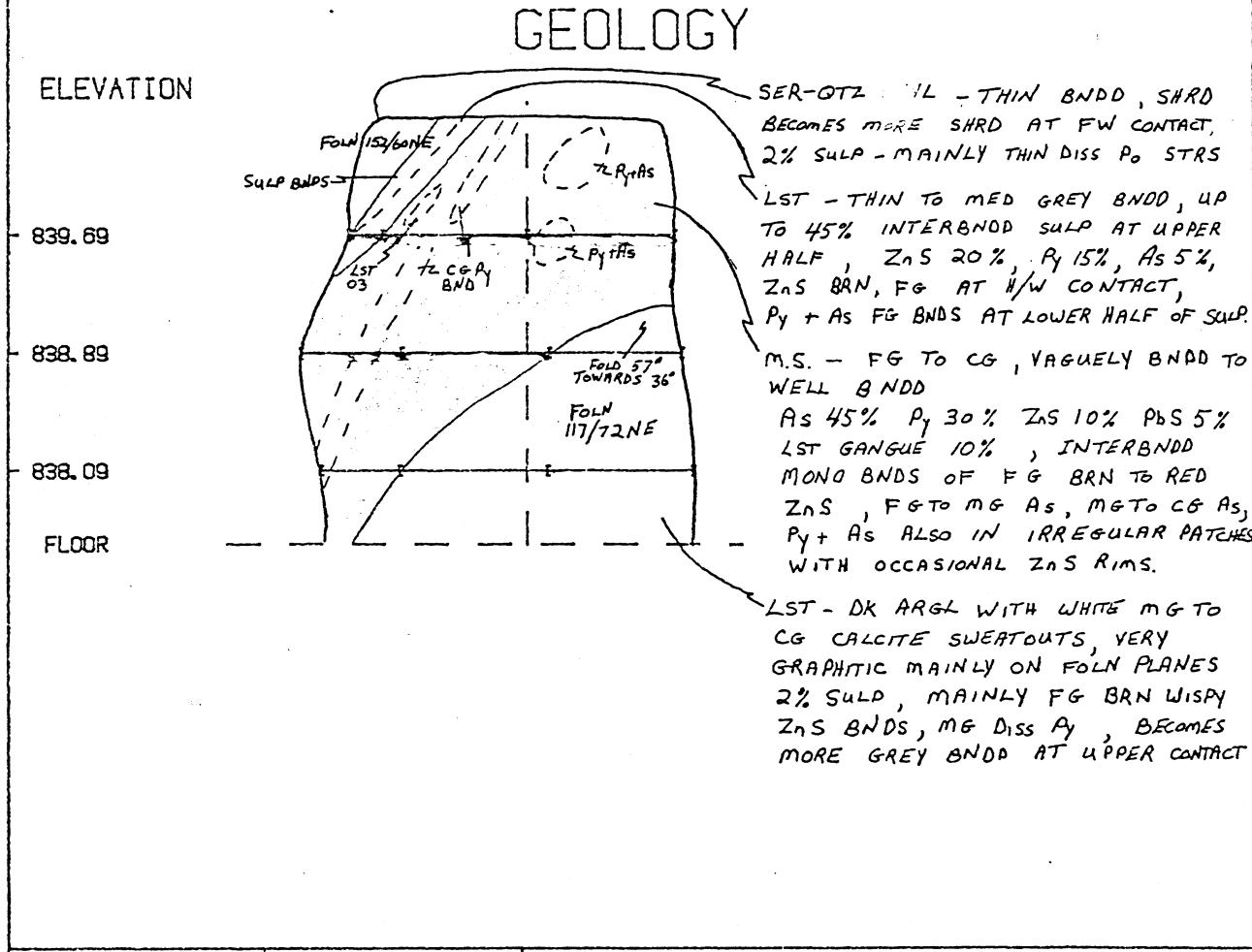
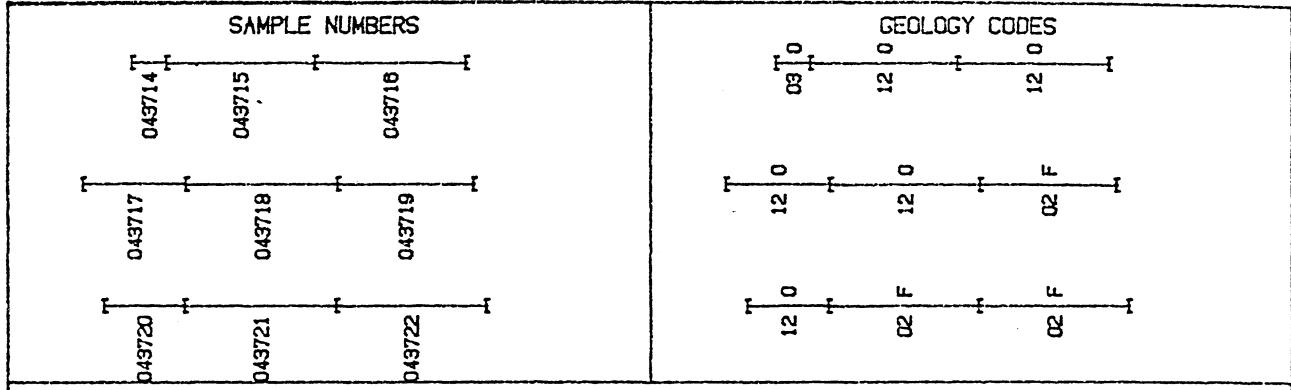
SELCO DIV. - BP RESOURCES

J & L PROJECT, B.C.

FACE# 152

* * * * *

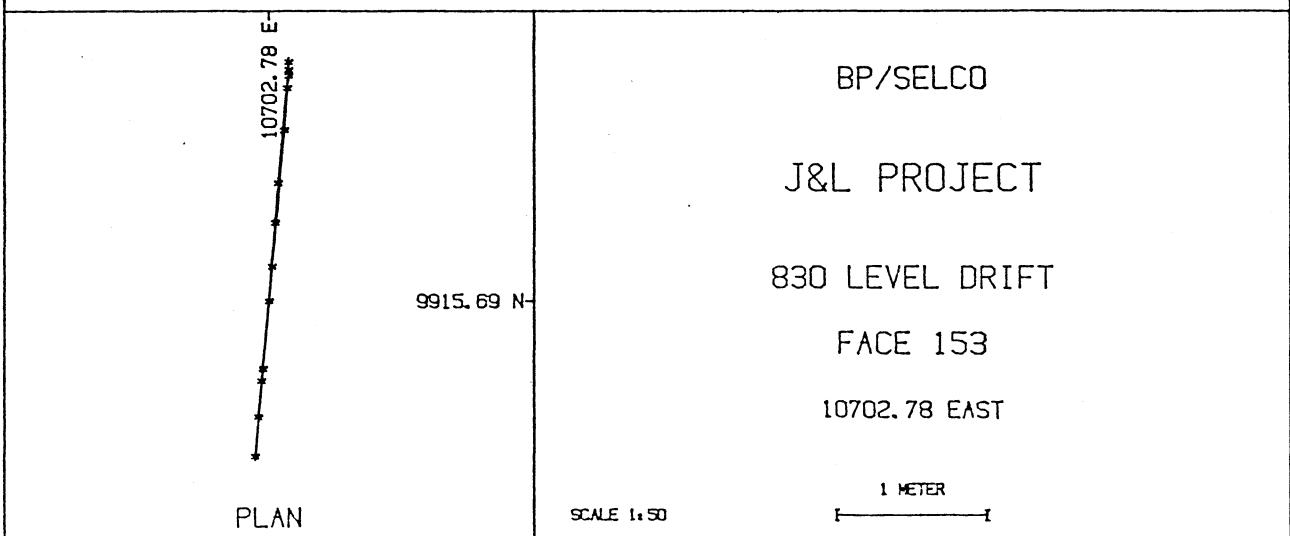
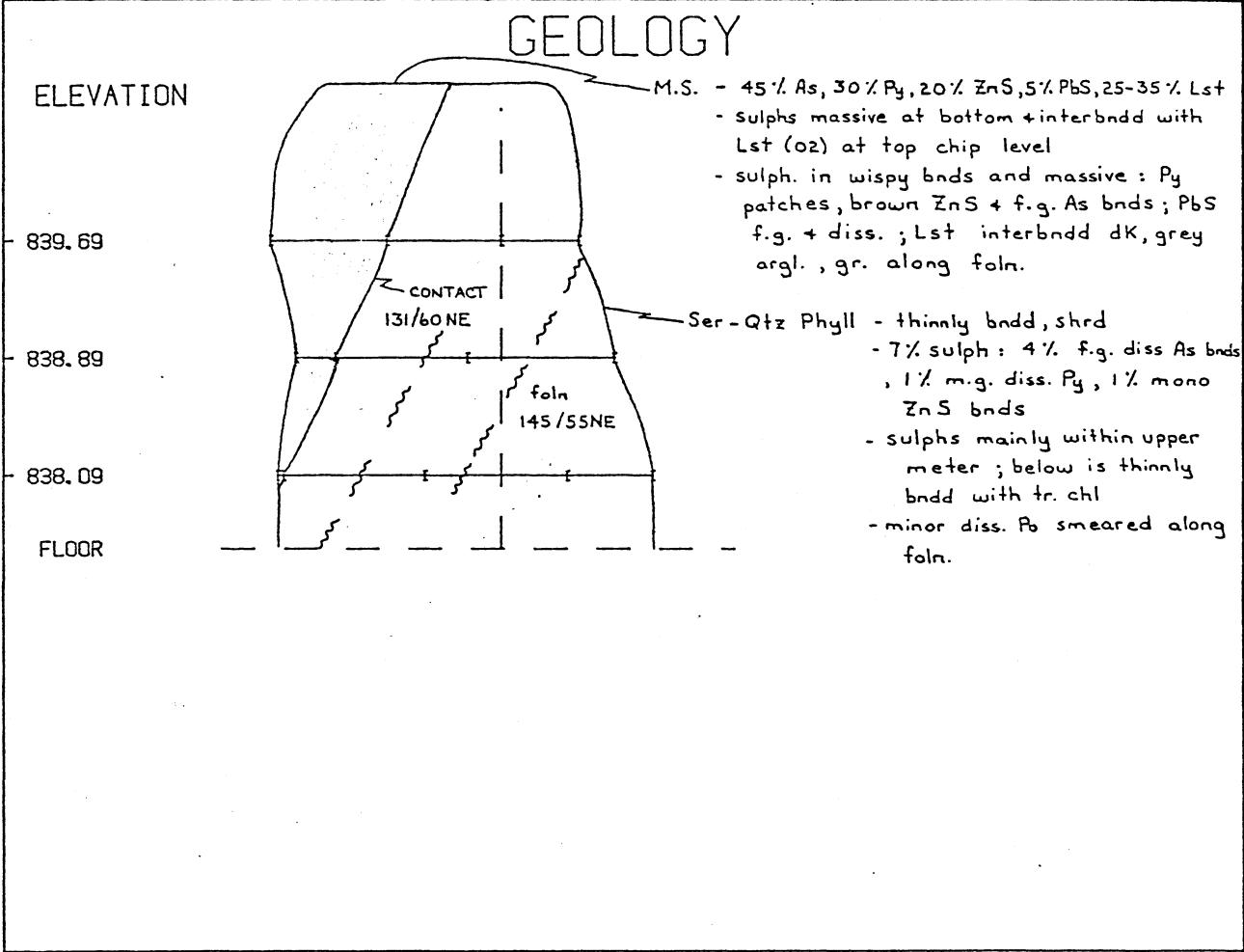
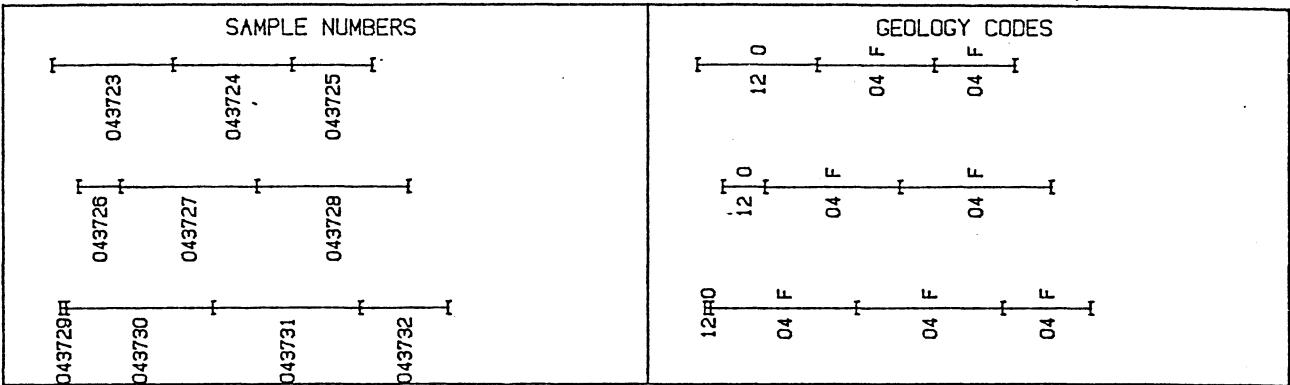
SAMPLE NUMBER	WIDTH (in)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43714	.23	2.86	3.30	4.440	54.2	6.8
43715	.98	.90	3.10	13.300	75.7	15.5
43716	1.00	.70	4.50	18.000	22.9	20.3
43717	.69	2.50	5.20	12.600	81.5	15.2
43718	1.00	.50	4.70	15.200	28.1	14.4
43719	.90	.05	.26	.528	.3	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 153

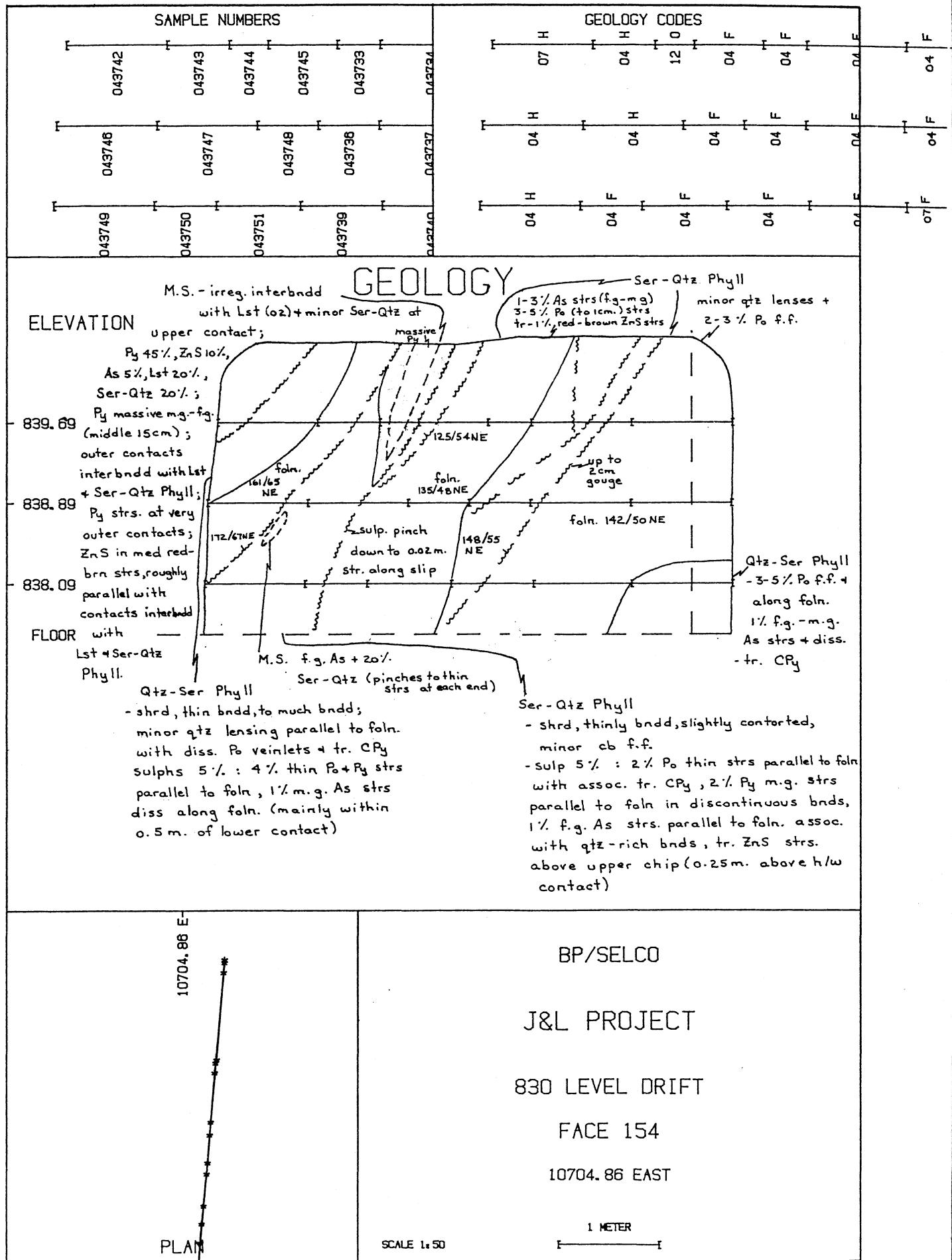
SAMPLE NUMBER	WIDTH (cm)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43723	.80	.90	2.40	8.080	30.8	4.2
43724	.78	.08	.04	2.190	1.3	.7
43725	.53	.02	.02	.753	2.4	.3
43726	.28	1.68	10.70	11.500	54.7	10.4
43727	.90	.05	.11	1.470	8.1	.8
43728	1.00	.01	.04	.094	2.6	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 154

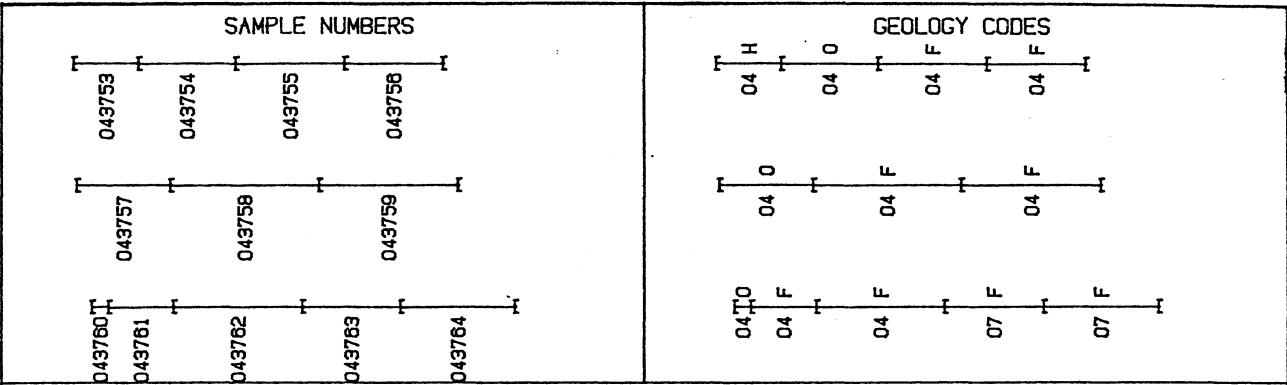
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43733	.43	.01	.01	.414	5.2	.3
43734	1.00	.01	.01	.325	2.1	.6
43735	1.00	.01	.01	.054	3.2	.2
43736	.61	.01	.01	.247	.3	.4
43737	1.00	.01	.01	.214	.4	.3
43738	1.00	.01	.01	.011	2.6	.1
43742	1.00	.01	.01	.054	3.3	.1
43743	.62	.06	.09	1.420	7.8	1.8
43744	.39	.32	2.25	1.660	20.5	2.0
43745	.69	.24	.07	4.540	7.2	3.8
43746	1.00	.01	.01	1.000	1.4	.3
43747	1.00	.03	.03	1.180	4.7	.8
43748	.61	.01	.01	1.010	2.0	.7



SELCO DIV. - BP RESOURCES J & I. PROJECT, B.C.

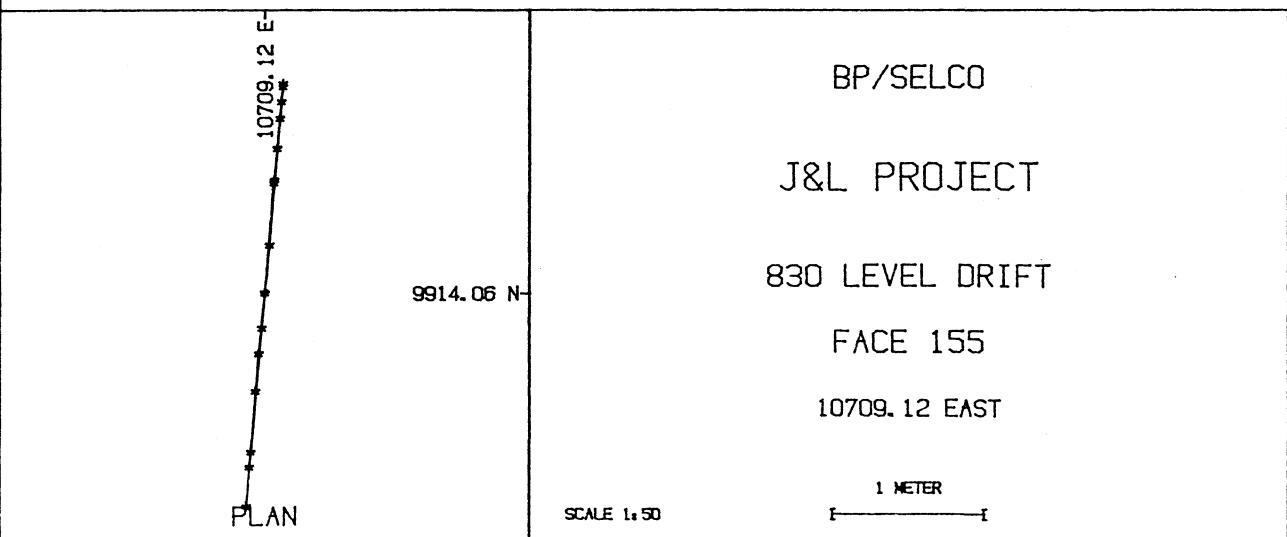
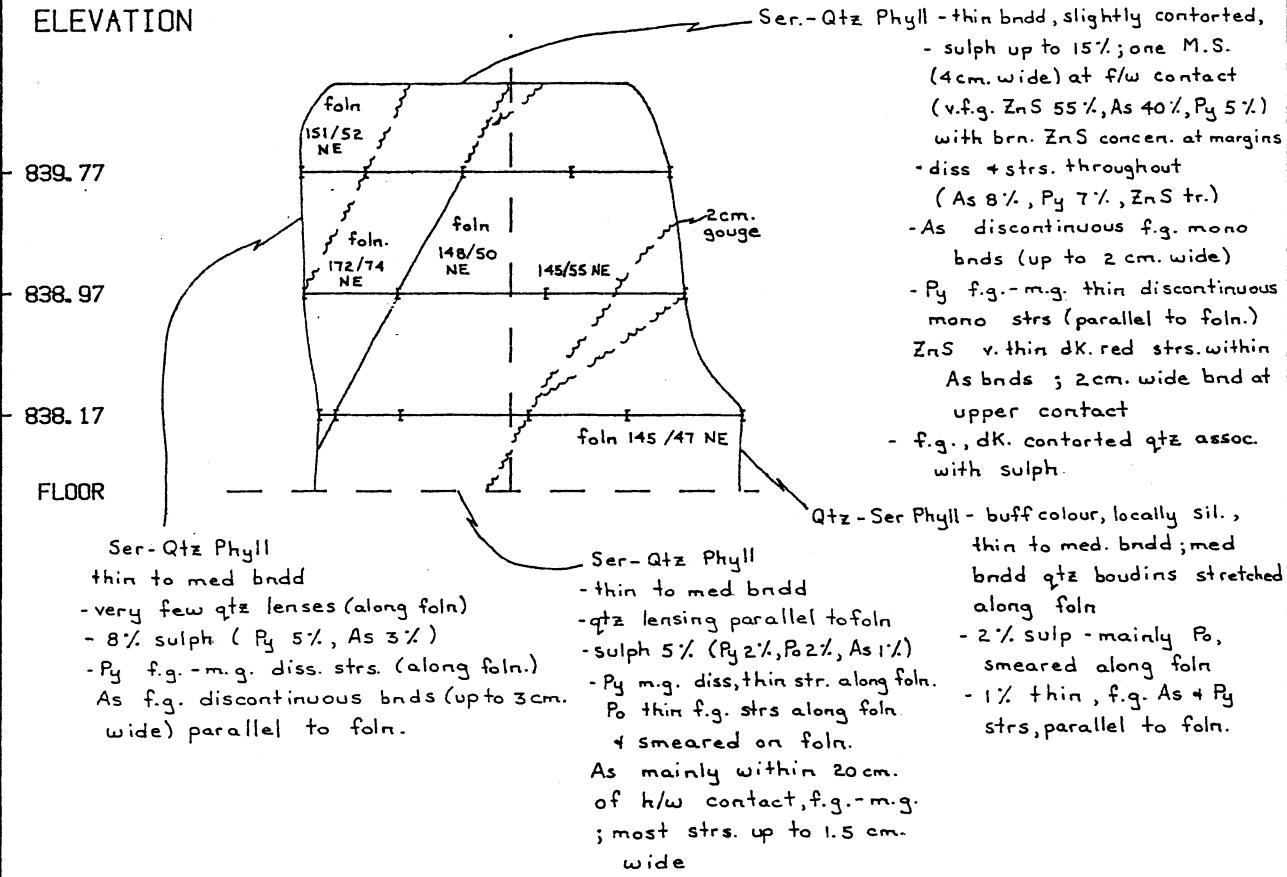
FACE# 155

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43753	.43	.24	.27	2.740	10.9	1.4
43754	.64	.1.17	1.22	2.880	25.5	3.3
43755	.72	.04	.02	1.930	.3	.6
43756	.65	.01	.01	.264	.3	.8
43757	.62	.65	1.02	4.240	19.1	3.5
43758	.98	.02	.01	1.090	2.4	.3
43759	.92	.01	.01	.140	16.7	.4



GEOLOGY

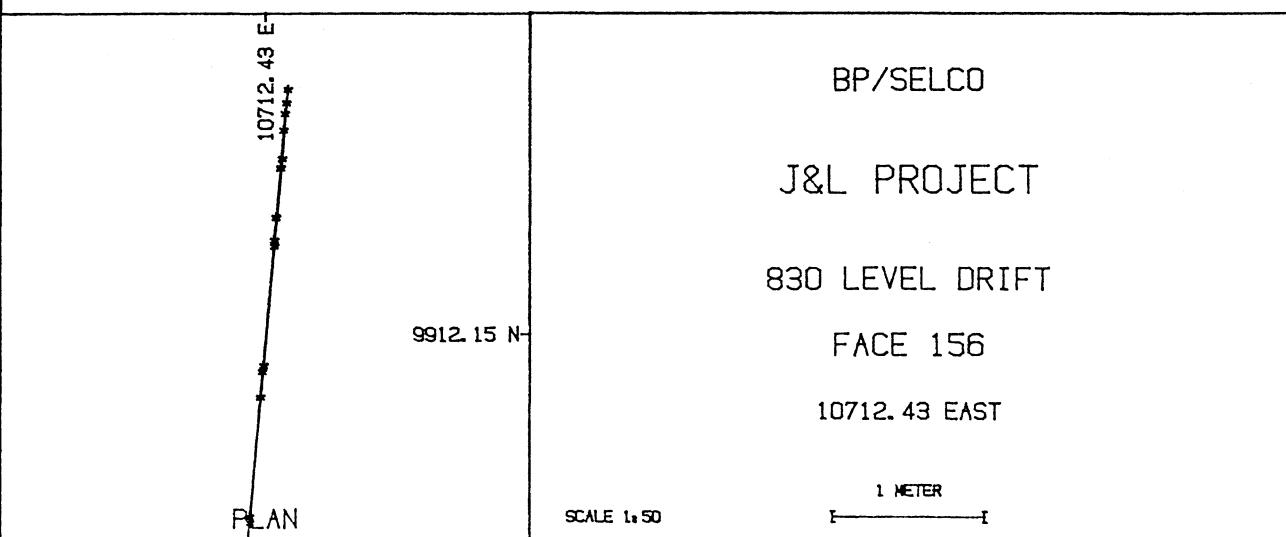
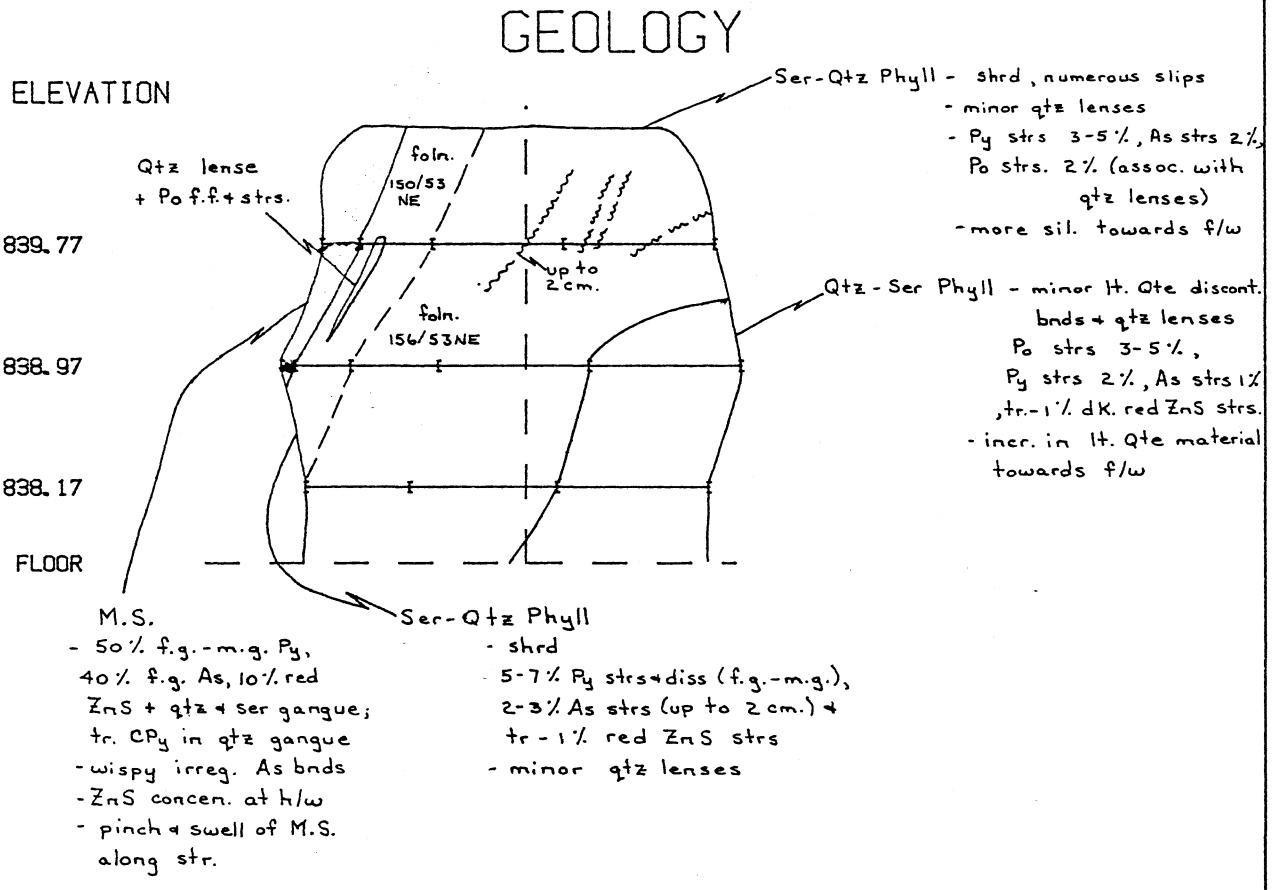
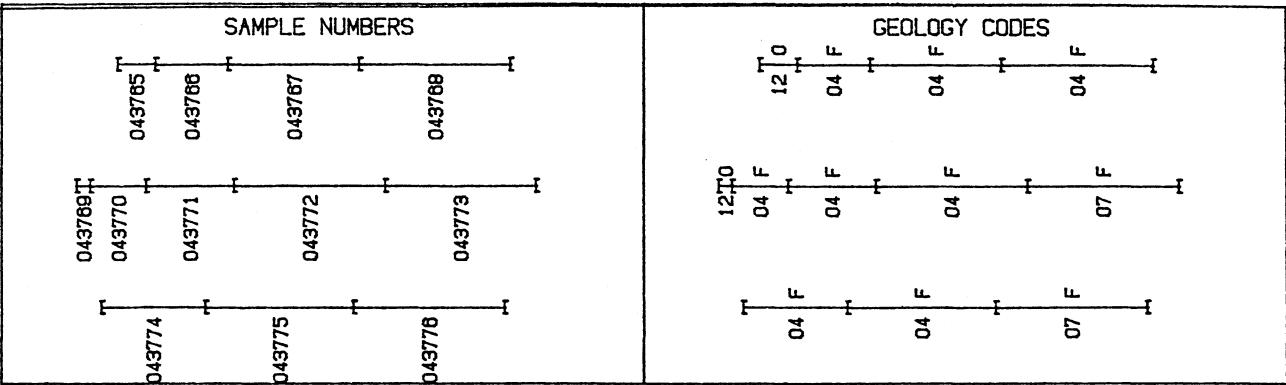
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 156

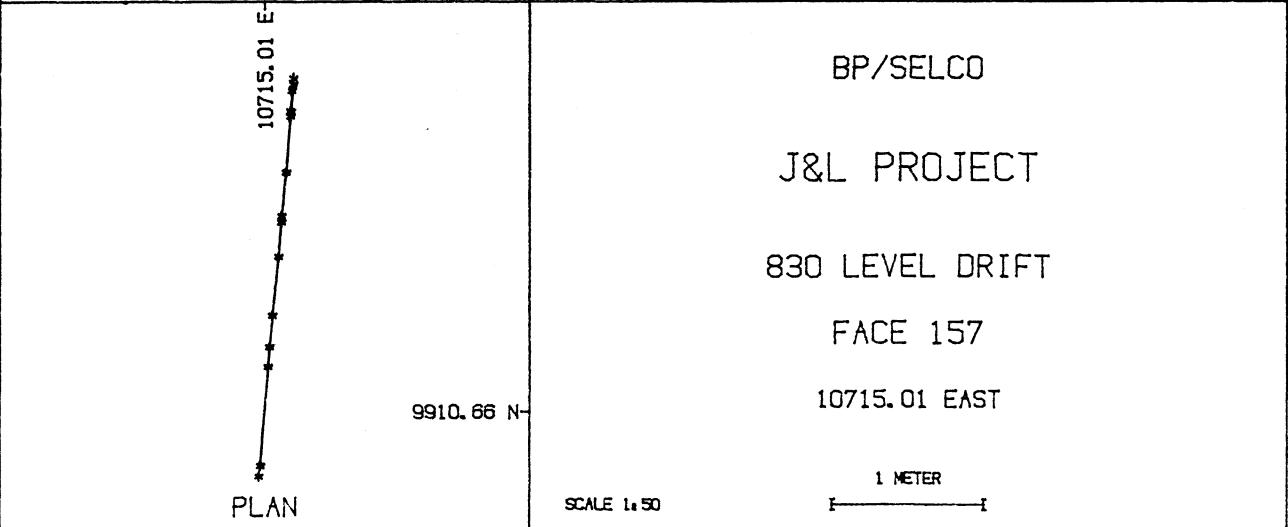
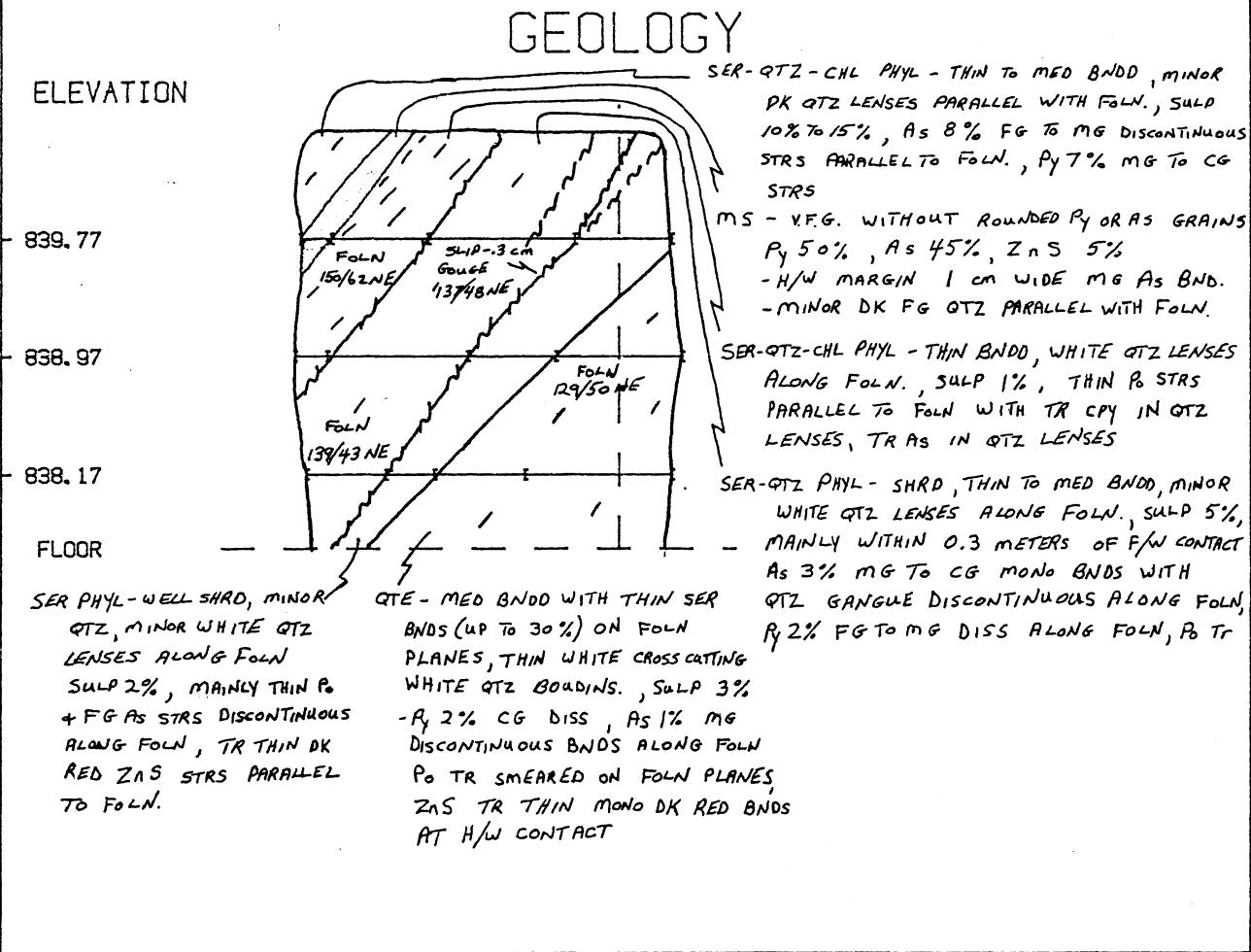
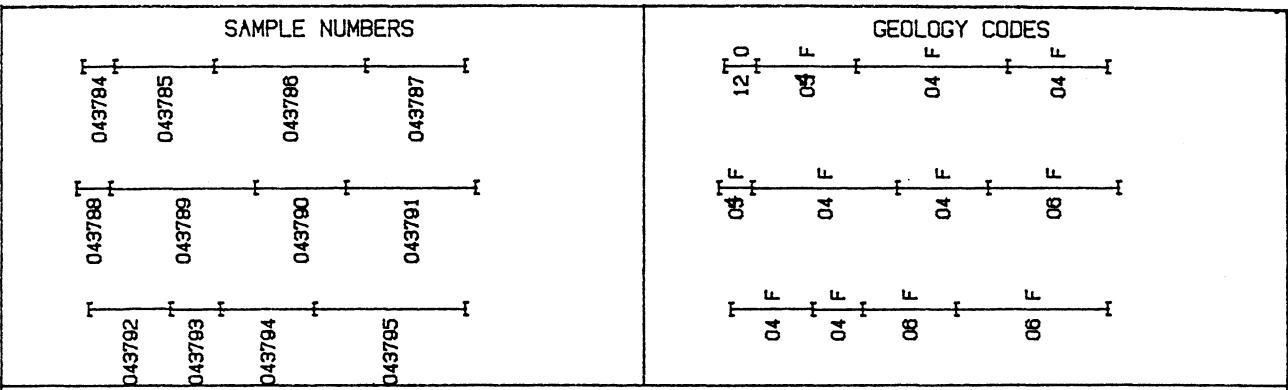
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43765	.25	.28	.15	13.700	61.9	19.7
43766	.48	.09	.06	1.790	.3	.4
43767	.87	.01	.01	.439	.3	.8
43768	1.00	.01	.01	.339	1.2	.2
43769	.09	5.94	10.70	7.460	145.9	13.1
43770	.37	.12	.16	1.990	2.7	.7
43771	.58	.01	.02	.359	.3	.6
43772	1.00	.02	.03	.851	.3	.3
43773	1.00	.03	.06	.744	1.0	.7



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 157

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43784	.21	1.33	3.36	9.090	43.8	9.0
43785	.66	.03	.03	.415	7.1	.4
43786	1.00	.05	.01	.993	10.5	.5
43787	.66	.11	.06	.438	7.1	.4
43788	.22	.03	.01	.972	2.4	.7
43789	.96	.01	.01	.807	5.8	.4
43790	.60	.04	.05	.262	10.7	.3
43791	.86	.01	.01	.148	8.0	.2

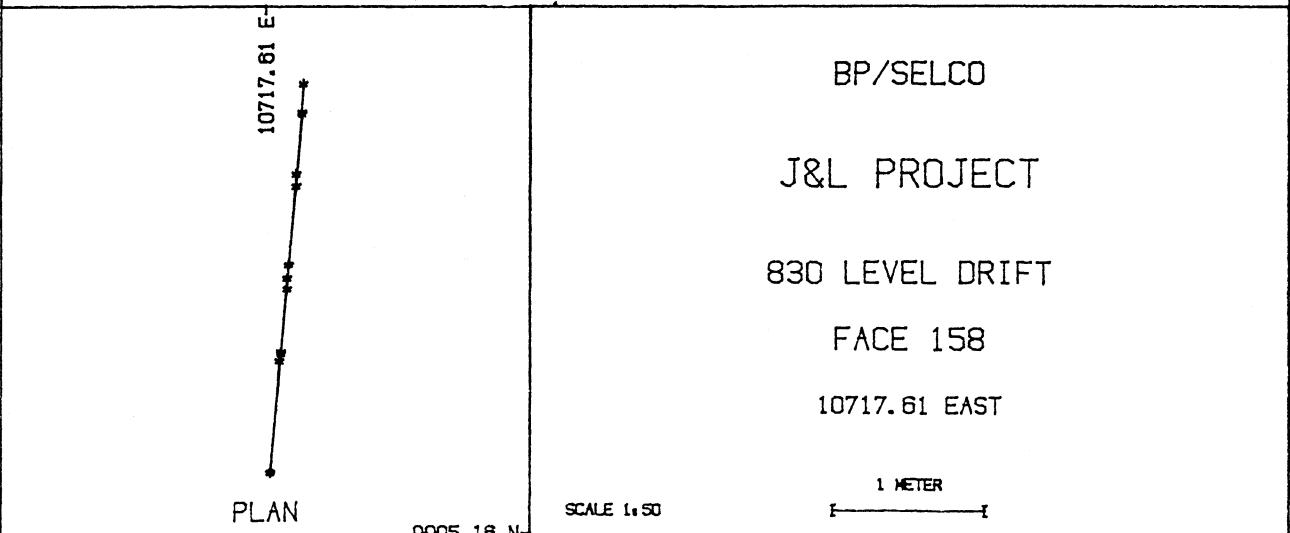
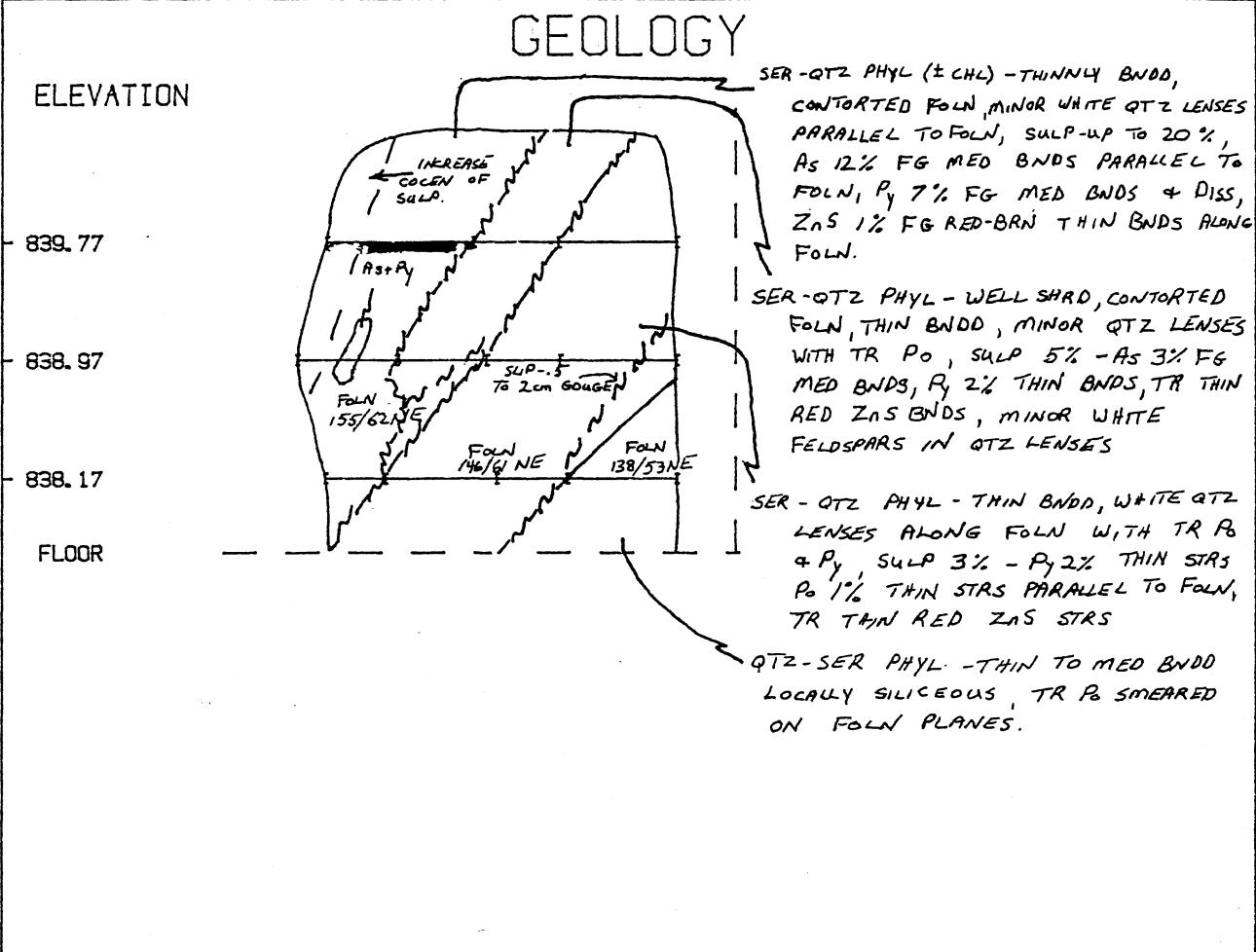


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 158

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43797	1.00	.39	.62	9.320	20.2	8.6
43798	.58	.22	.24	2.260	10.2	2.1
43799	.80	.02	.02	.319	5.9	.3
43800	.68	.16	.12	6.340	12.1	3.0
43801	.61	.48	1.08	3.290	16.0	3.9
43802	.50	.04	.05	.500	5.1	.4
43803	.79	.03	.04	1.020	5.7	.5

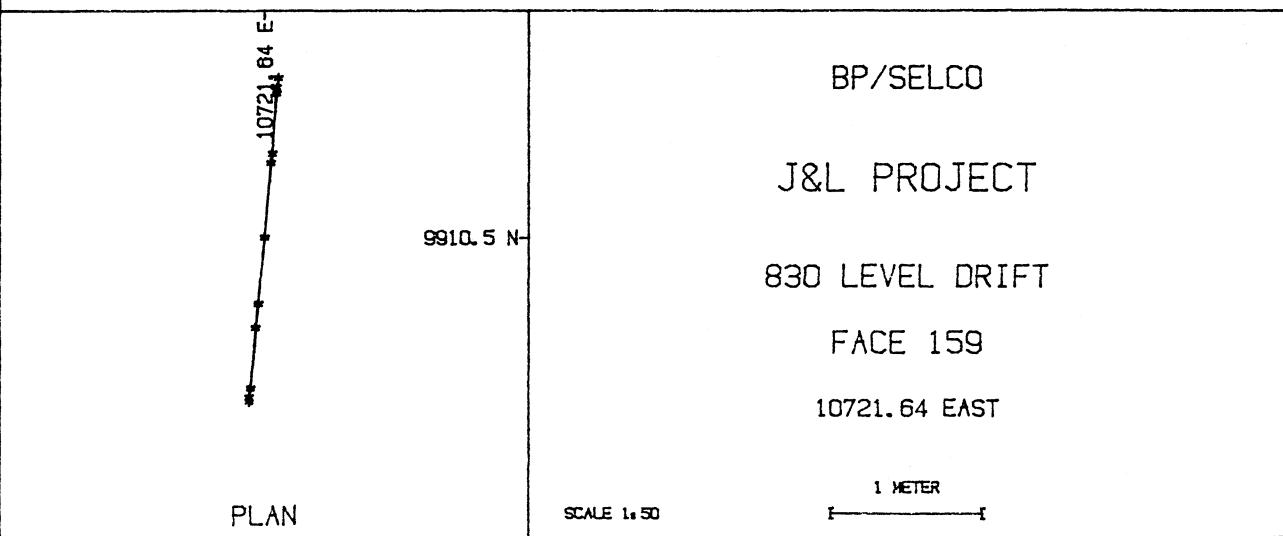
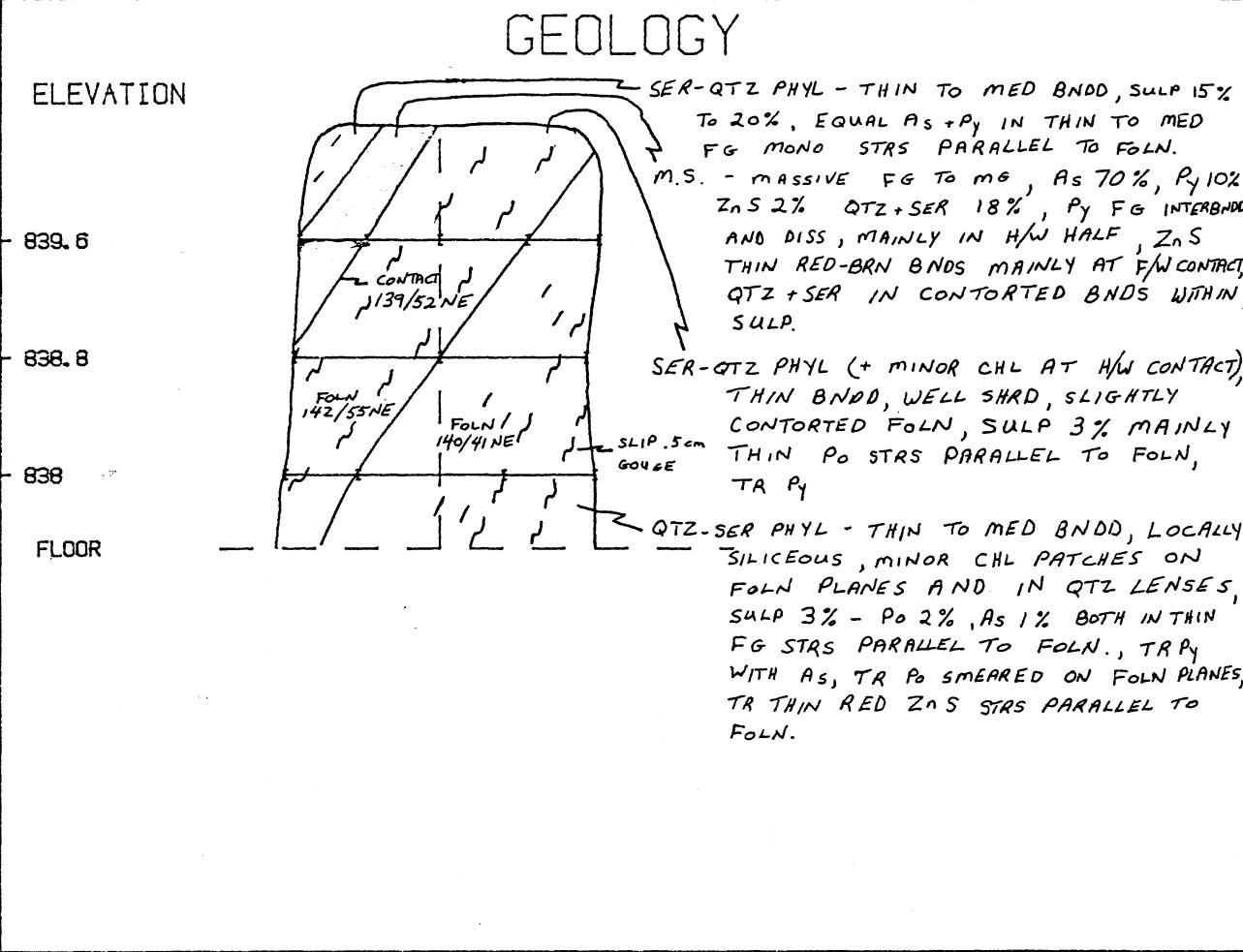
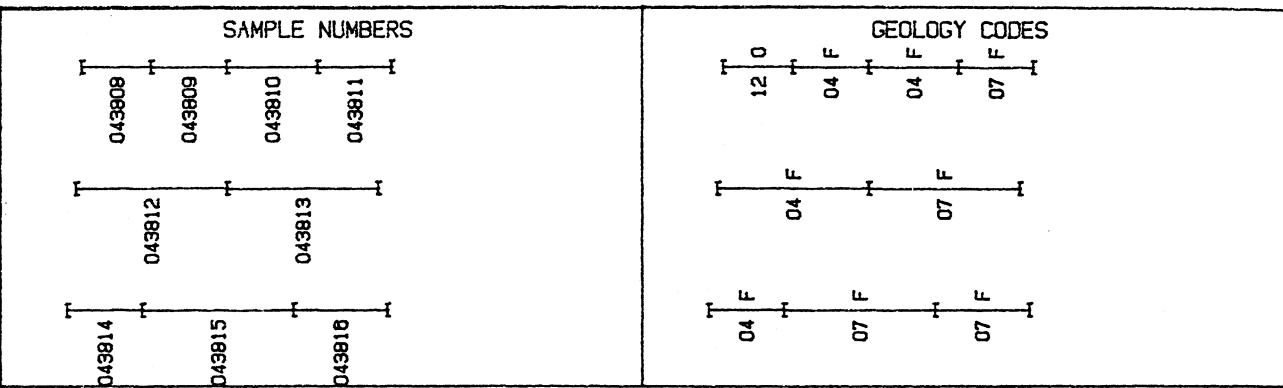
SAMPLE NUMBERS			GEOLOGY CODES		
043787	043798	043799	04 F	04 F	04 F
043800	043801	043802	04 F	04 F	04 F
043804	043805	043806	04 F	04 F	04 F
043807					07 F



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 159

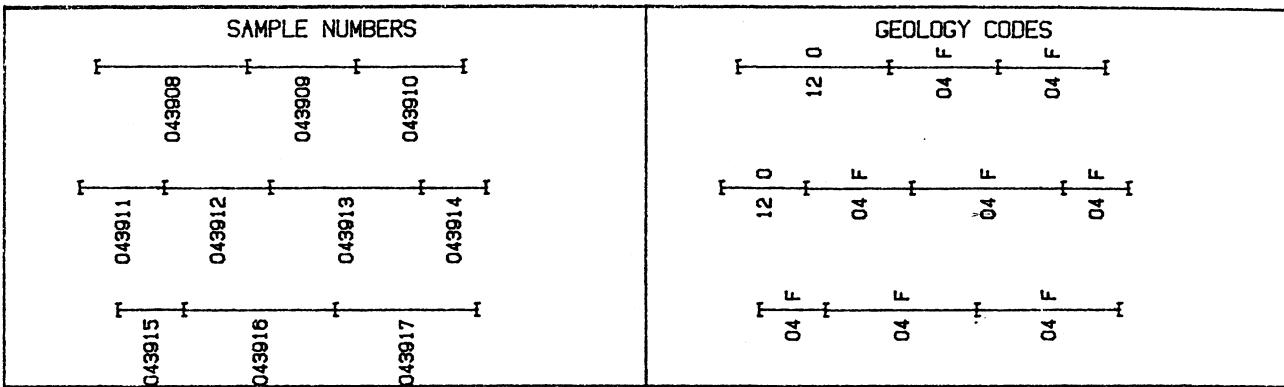
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43808	.46	.77	.82	12.900	45.0	8.5
43809	.50	.03	.01	1.160	4.8	.7
43810	.60	.01	.01	.381	.4	.3
43811	.49	.01	.01	.468	.3	.5
43812	1.00	.01	.01	.081	.4	.3
43813	1.00	.01	.01	.332	.3	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

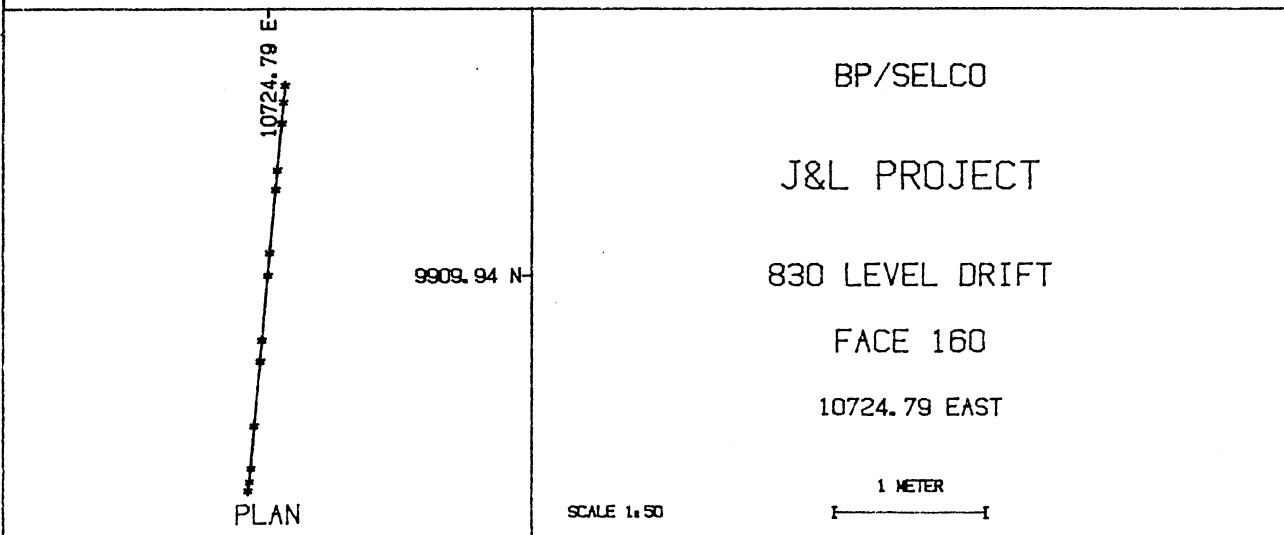
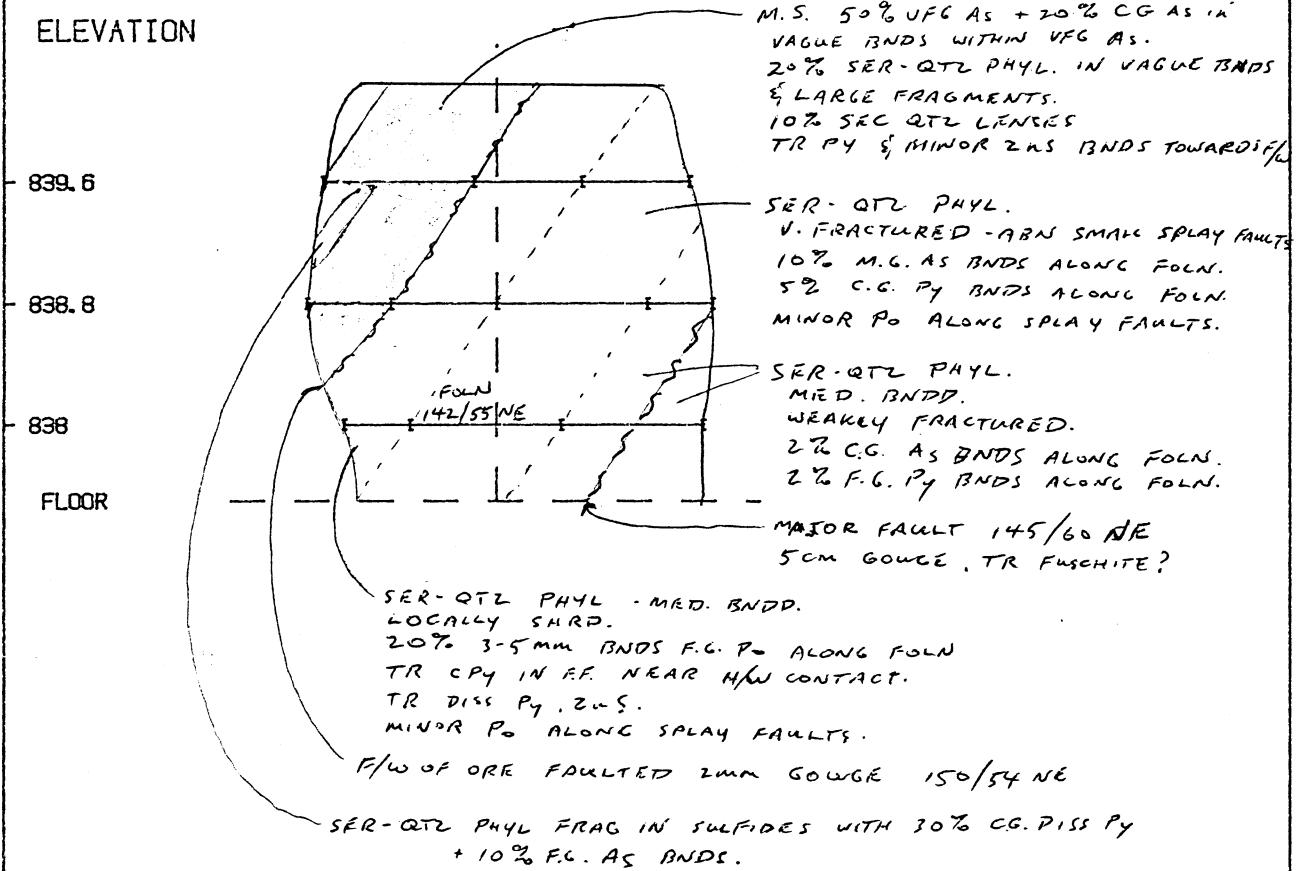
FACE# 160

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43908	1.00	.29	.10	13.700	12.7	4.4
43909	.72	.03	.01	1.160	.3	.3
43910	.71	.01	.01	.931	.3	.4
43911	.56	.23	.02	18.300	9.0	4.7
43912	.70	.02	.01	.952	2.9	.5
43913	1.00	.01	.01	.909	2.3	.4
43914	.43	.01	.01	.172	.3	.3



GEOLOGY

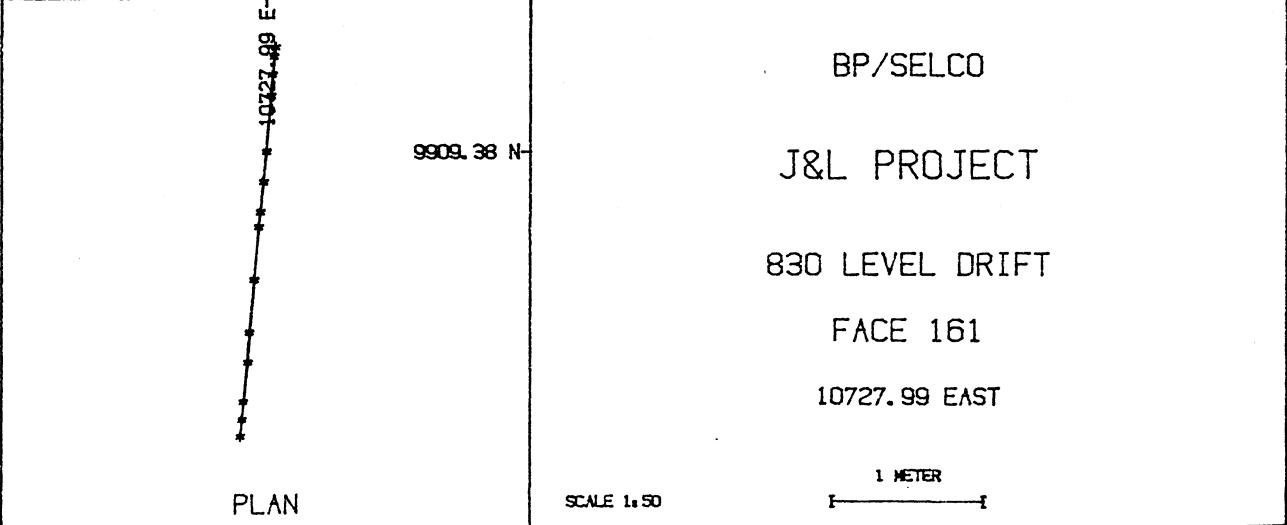
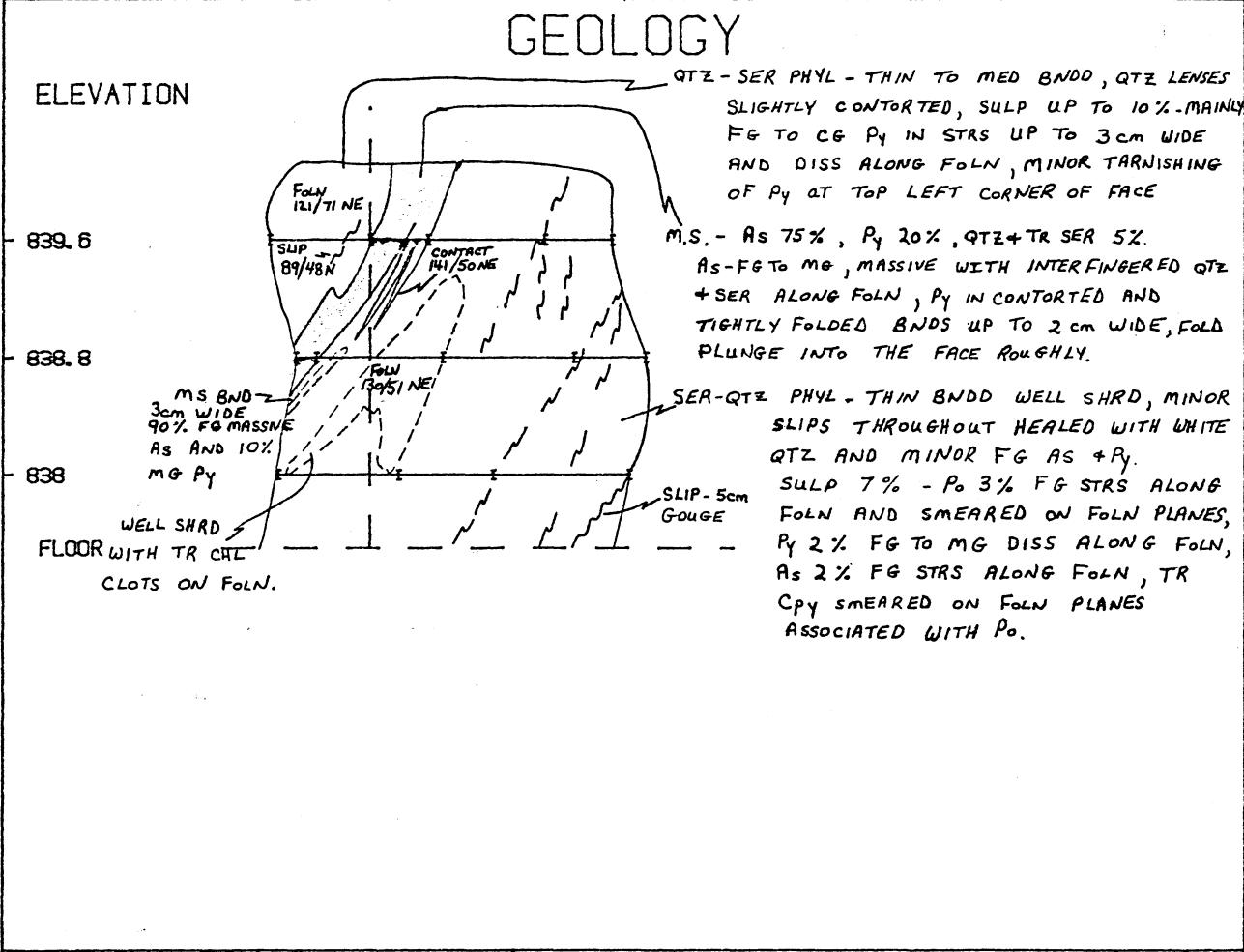
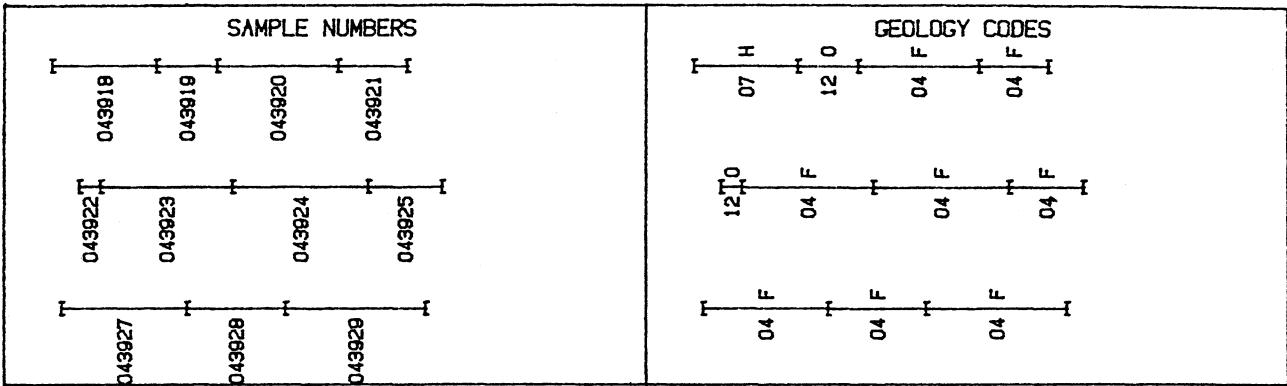
ELEVATION



SELCO DIV. - BF RESOURCES J & L PROJECT, B.C.

FACE# 161

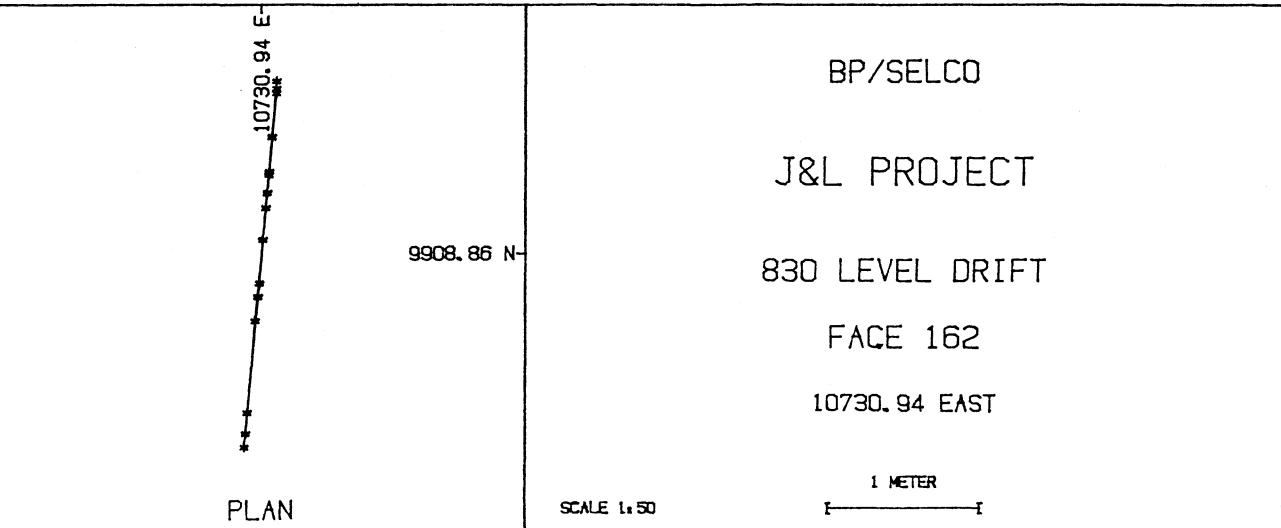
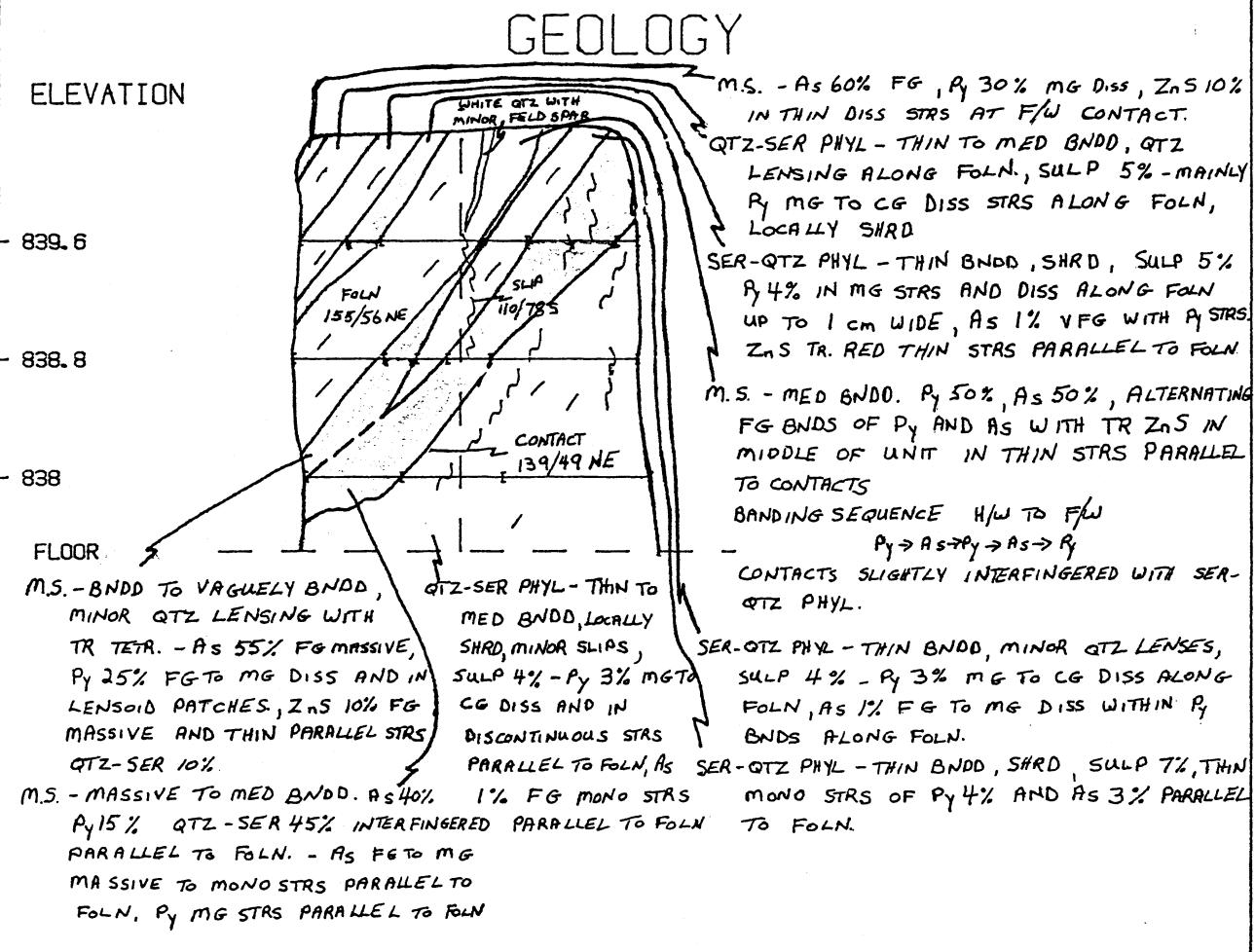
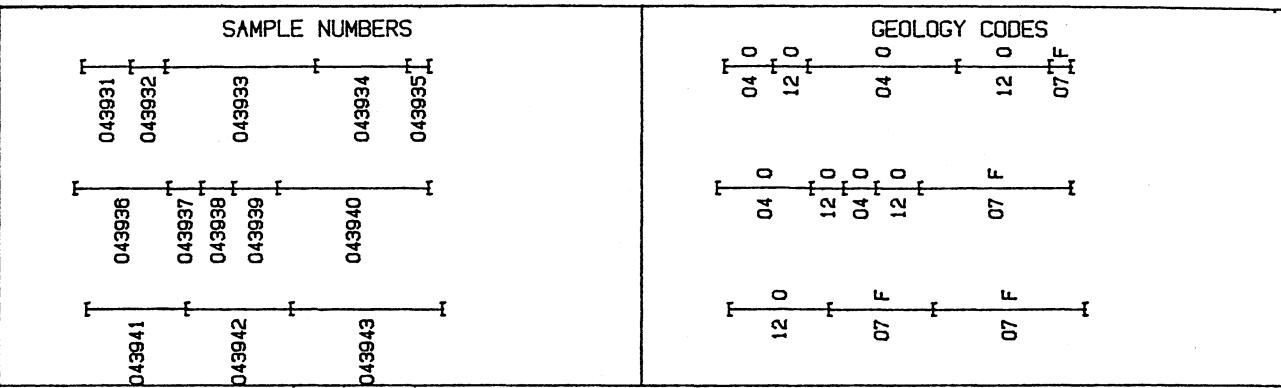
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43918	.64	.07	.01	2.170	2.0	1.4
43919	.40	.19	.01	16.800	14.3	13.1
43920	.80	.01	.01	1.210	2.0	.7
43921	.46	.02	.01	3.150	1.7	1.0
43922	.14	.33	.01	20.100	14.1	11.9
43923	.87	.04	.01	.695	1.0	.4
43924	.90	.01	.01	.815	1.7	.3
43925	.49	.01	.01	1.750	.3	.5



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 162

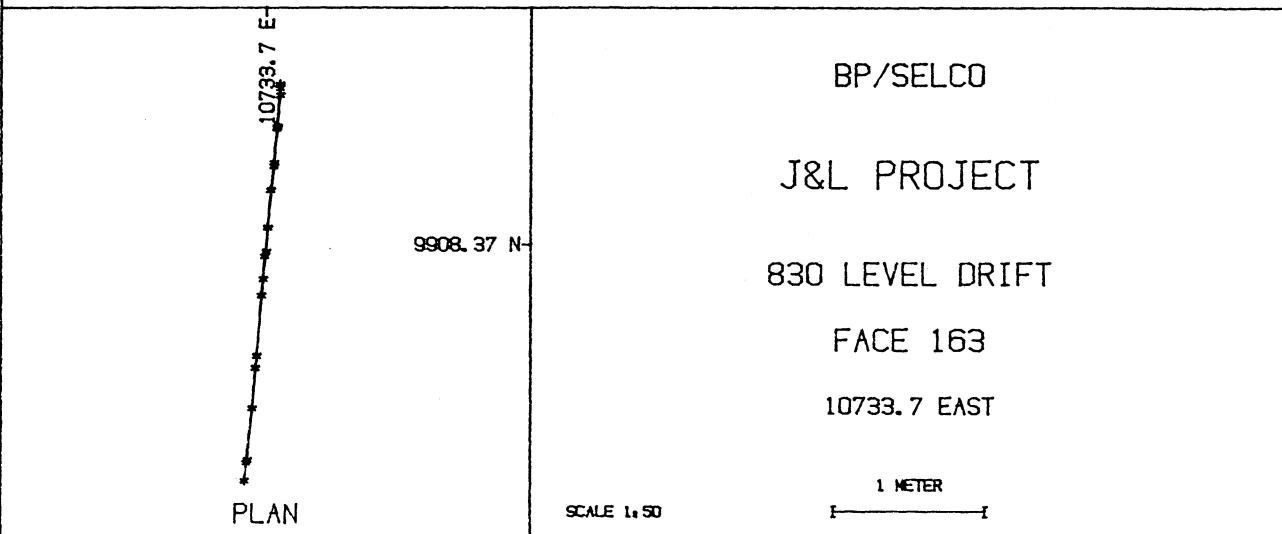
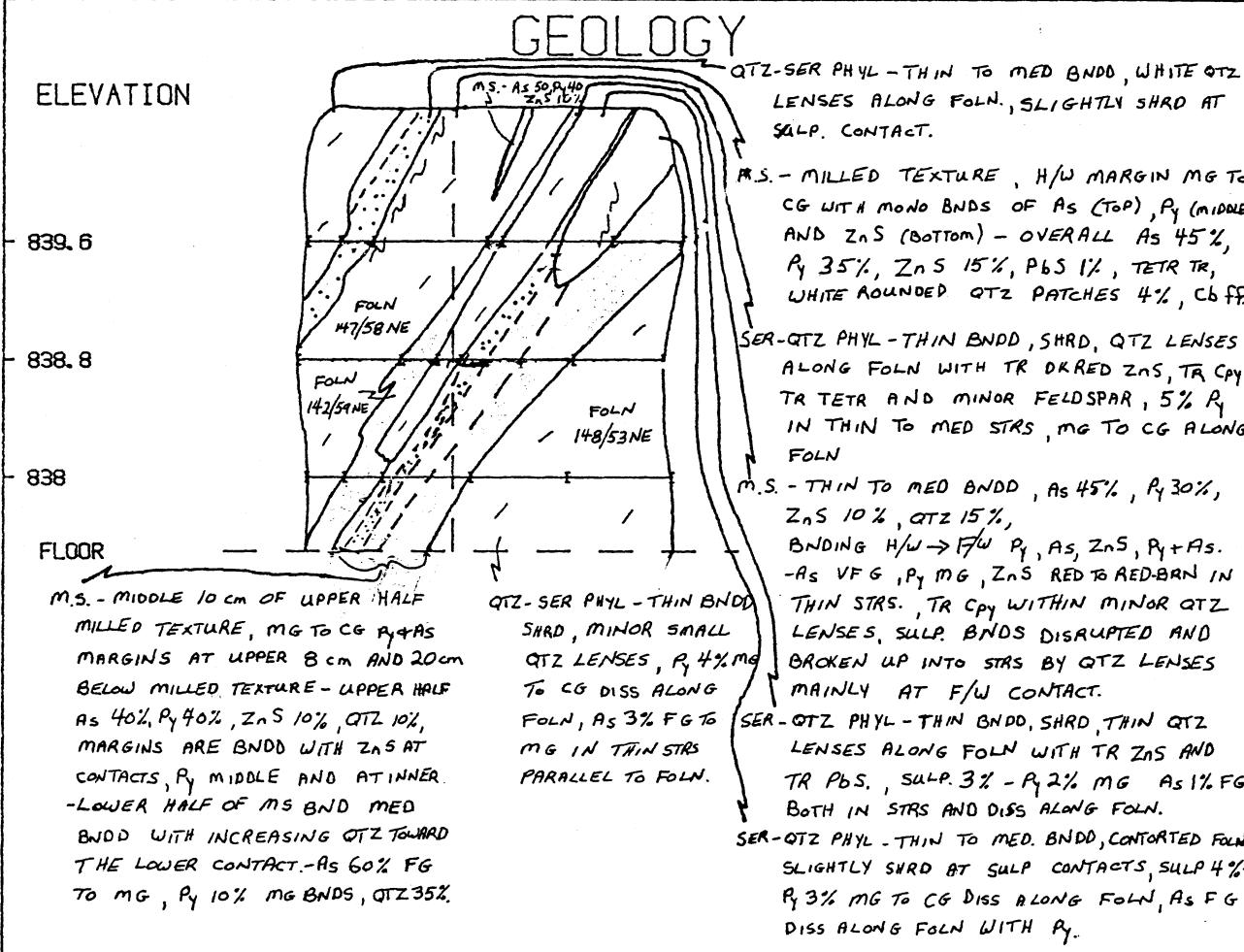
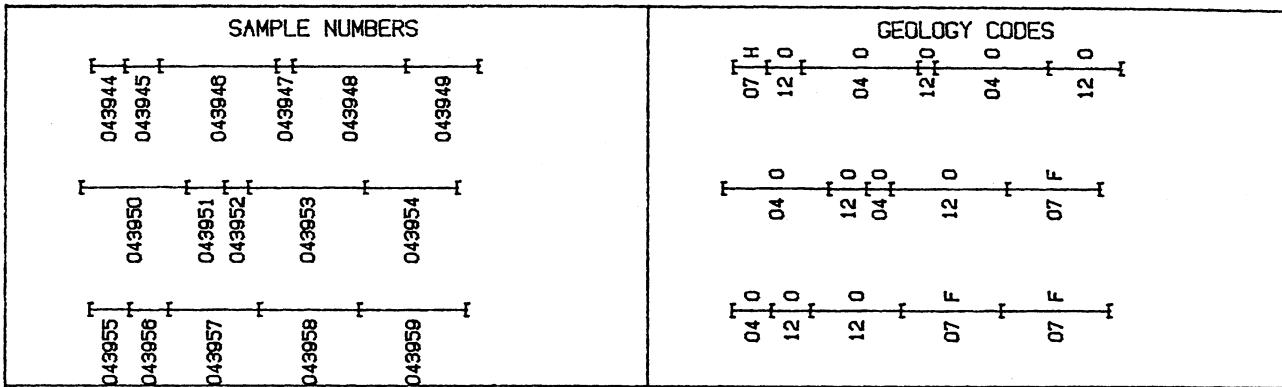
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43931	.32	.14	.03	.481	5.9	.3
43932	.23	1.84	.73	14.200	81.7	14.3
43933	.99	.45	.66	1.770	14.7	1.1
43934	.61	.21	.01	18.300	20.9	13.4
43935	.14	.02	.01	2.650	2.0	.7
43936	.62	.03	.03	7.030	.3	.6
43937	.22	3.68	8.22	6.450	72.2	6.6
43938	.21	.16	.20	1.440	4.3	.5
43939	.29	.16	.02	14.900	8.3	2.0
43940	1.00	.03	.01	2.490	4.8	.7



SELCO DIV. - BF RESOURCES J & L PROJECT, B.C.

FACE# 163

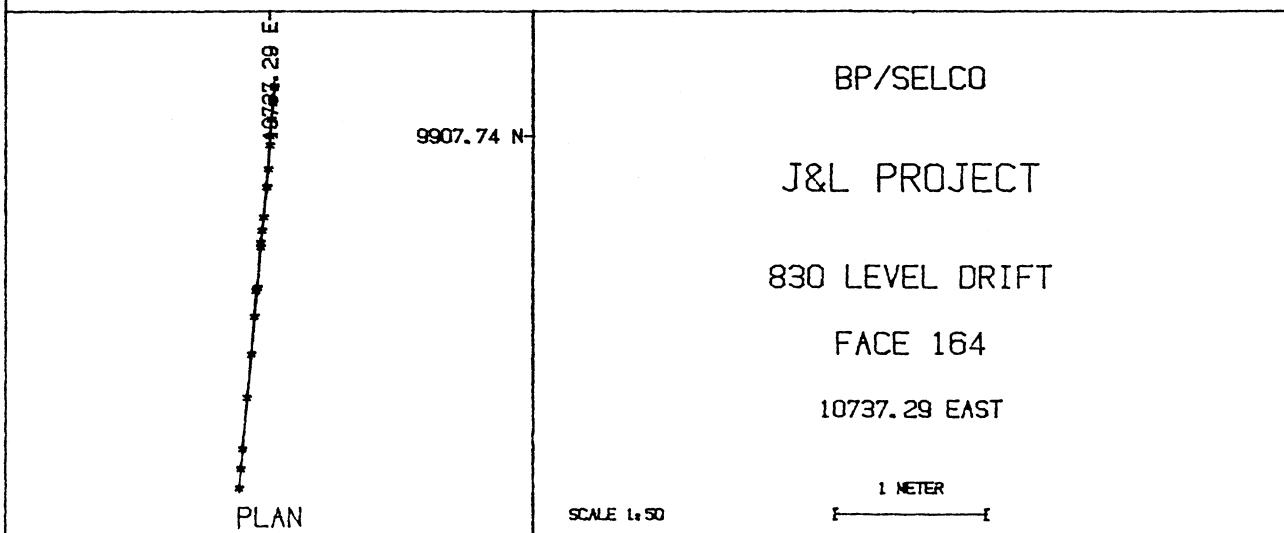
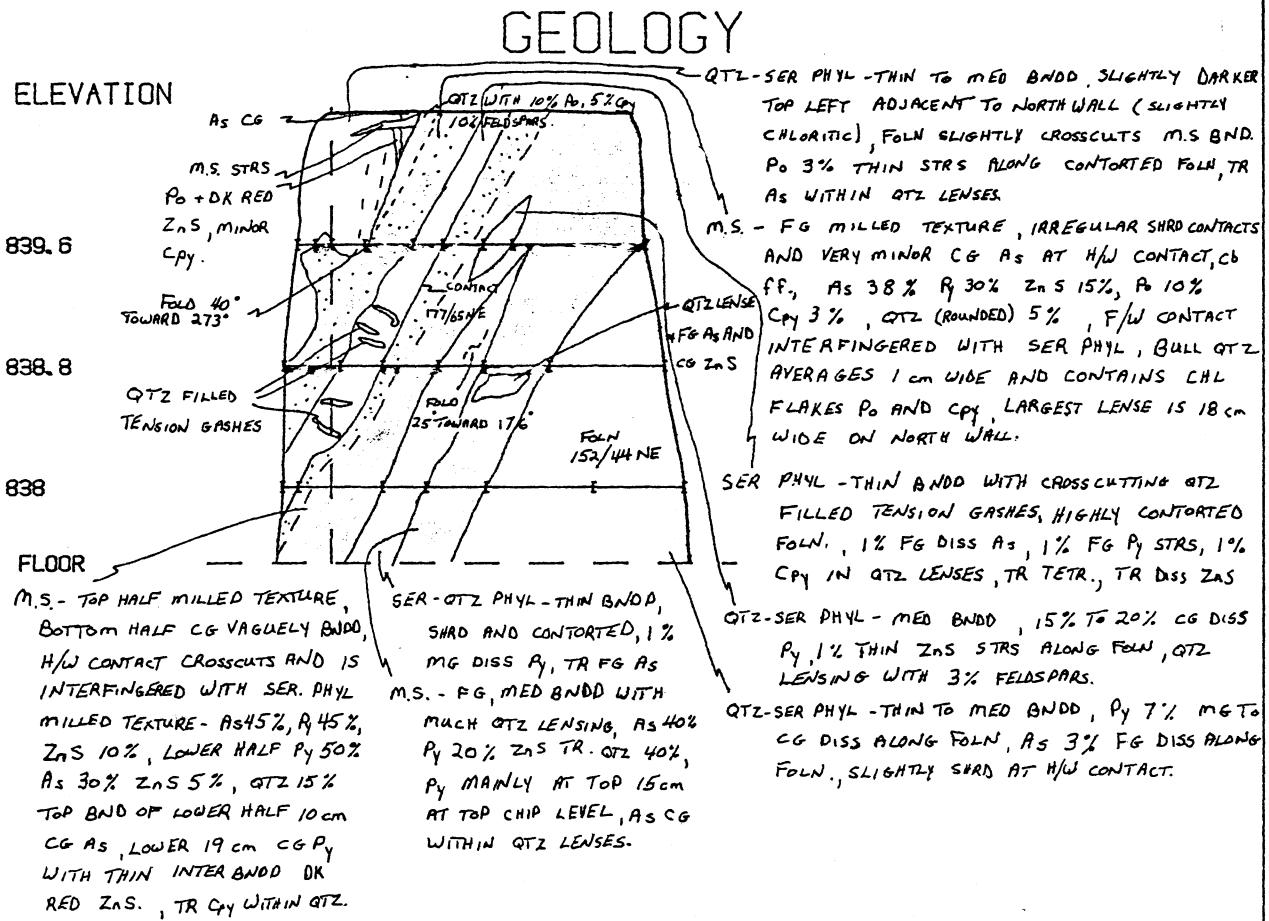
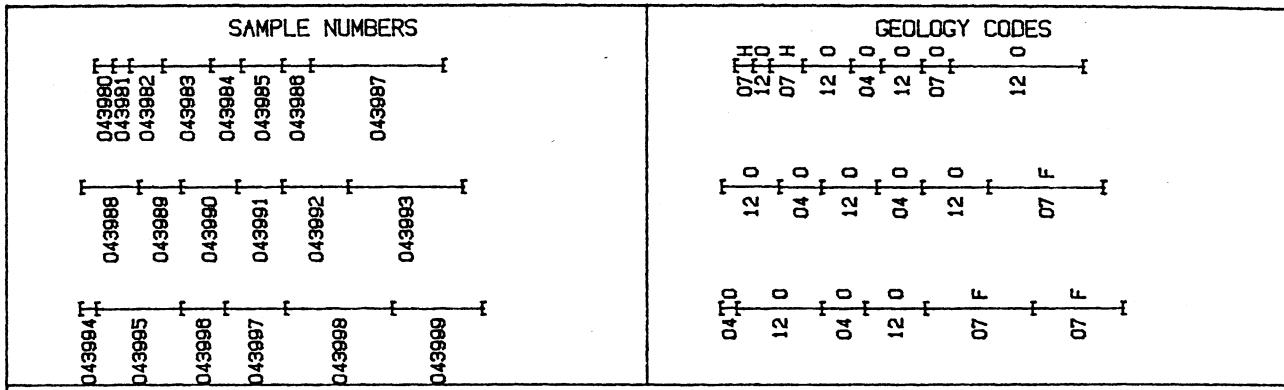
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43944	.22	.32	1.11	.284	16.5	.6
43945	.23	11.60	17.60	2.880	206.8	11.2
43946	.77	.20	.15	.188	7.2	.3
43947	.11	.82	1.46	11.100	26.6	10.4
43948	.75	1.24	1.67	3.970	25.7	3.0
43949	.48	.20	.11	7.230	7.4	4.9
43950	.70	.25	.13	7.200	8.5	4.5
43951	.25	1.17	1.43	11.800	42.8	9.9
43952	.16	.10	.04	2.040	6.1	1.4
43953	.77	4.06	8.22	9.410	80.3	8.8
43954	.69	.13	.17	5.430	7.8	3.8



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 164

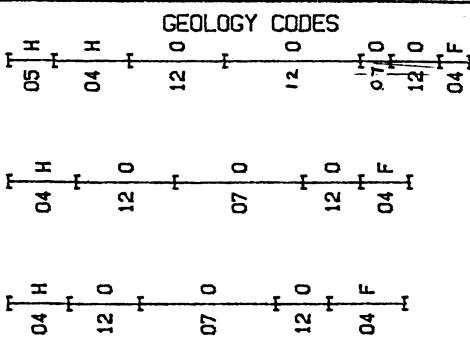
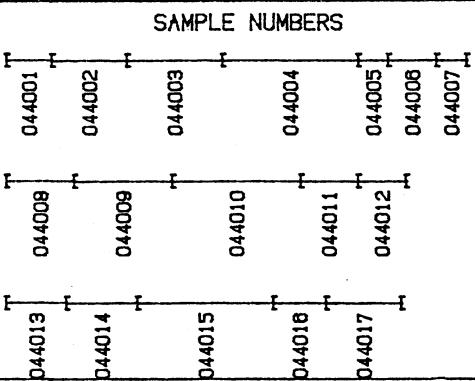
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
43980	.12	.19	.04	3.550	5.7	.4
43981	.11	4.06	9.20	6.180	39.1	12.3
43982	.22	.12	.07	3.090	6.5	1.0
43983	.32	10.90	13.70	3.280	177.5	9.9
43984	.20	2.34	2.96	1.140	46.6	2.0
43985	.27	8.57	15.80	7.650	132.7	15.3
43986	.19	.34	.35	1.810	9.4	.8
43987	.88	.57	.46	17.600	16.0	7.3
43988	.38	7.56	10.70	3.220	127.0	8.0
43989	.28	3.20	4.61	1.170	71.0	3.0
43990	.37	3.29	5.47	9.630	76.3	14.1
43991	.30	.30	.35	4.950	12.2	1.5
43992	.44	.37	.10	14.200	17.0	10.4
43993	.76	.05	.03	2.090	4.4	1.0



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

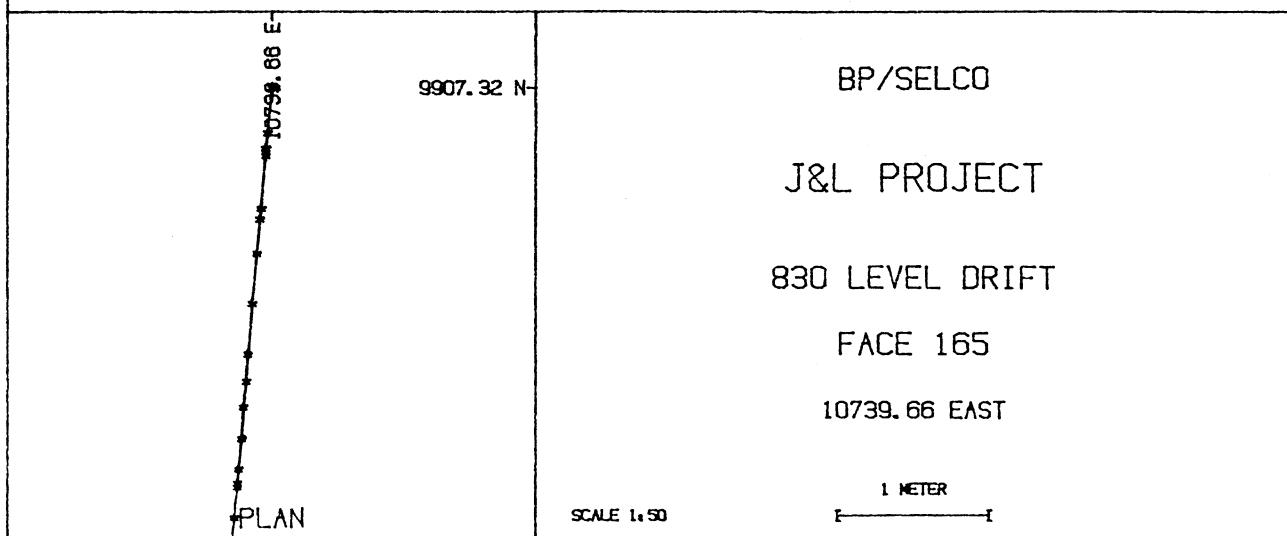
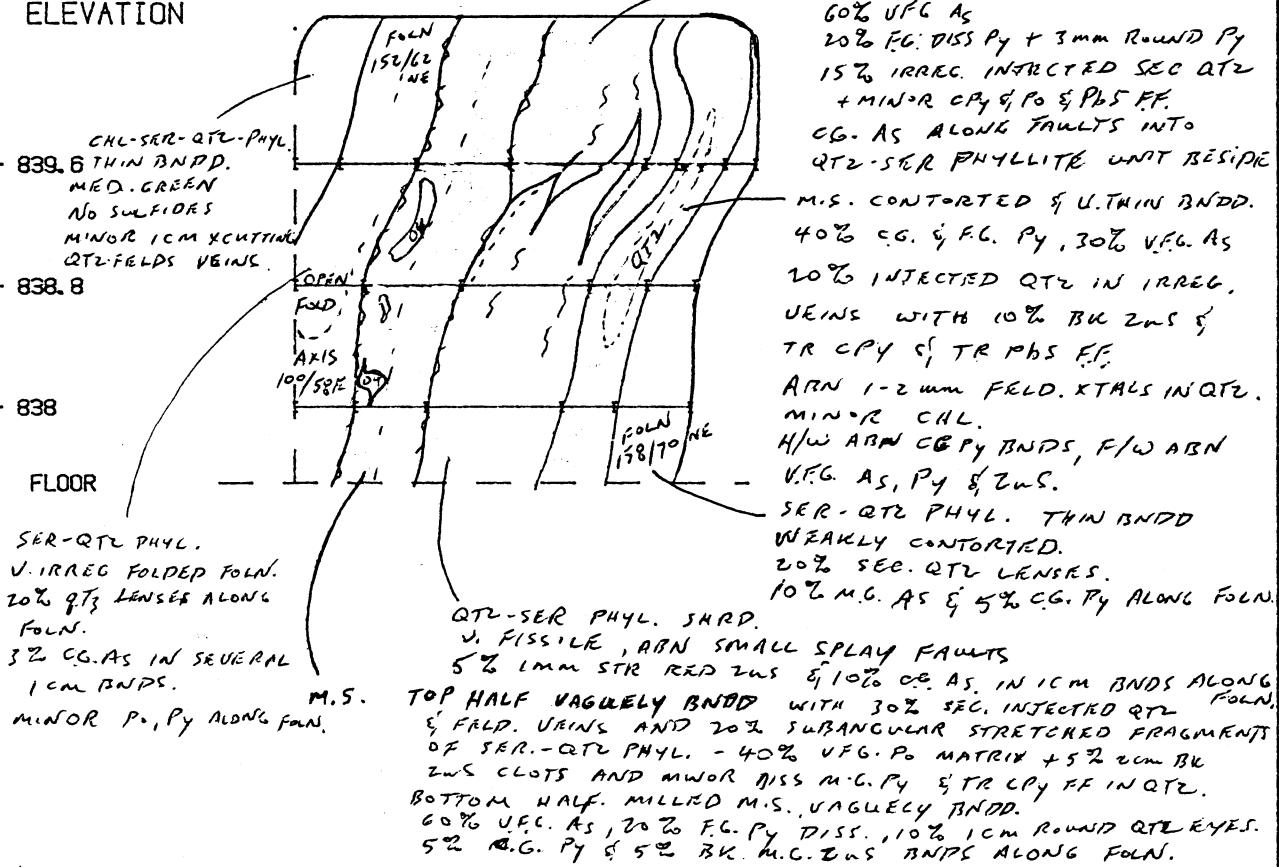
FACE# 165

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44001	.30	.01	.02	.045	.3	.2
44002	.50	.01	.07	.115	.3	.3
44003	.63	6.98	9.60	2.990	119.2	7.0
44004	.90	2.00	3.21	9.350	42.9	10.3
44005	.20	1.83	4.26	7.520	35.5	4.3
44006	.32	1.68	1.13	11.300	42.6	33.0
44007	.20	.08	.03	11.100	5.3	.9
44008	.45	.06	.13	1.210	1.7	.4
44009	.65	4.12	6.83	6.120	83.7	12.3
44010	.85	.79	.85	2.440	16.1	1.0
44011	.38	4.12	8.22	4.730	113.1	9.1
44012	.32	.21	.10	4.020	6.7	.8



GEOLOGY

ELEVATION

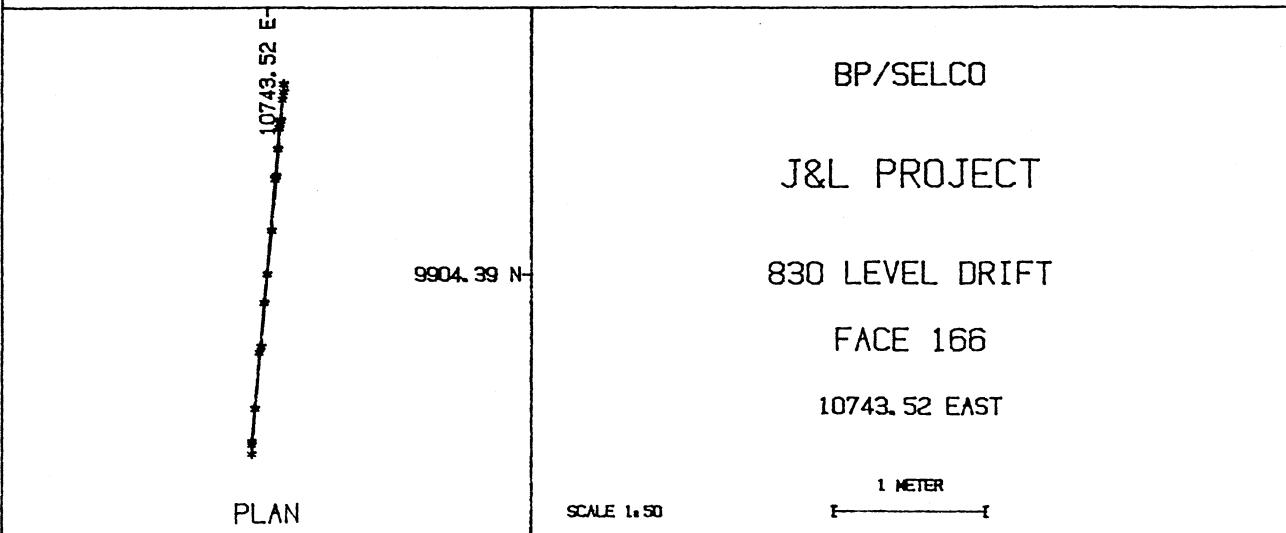
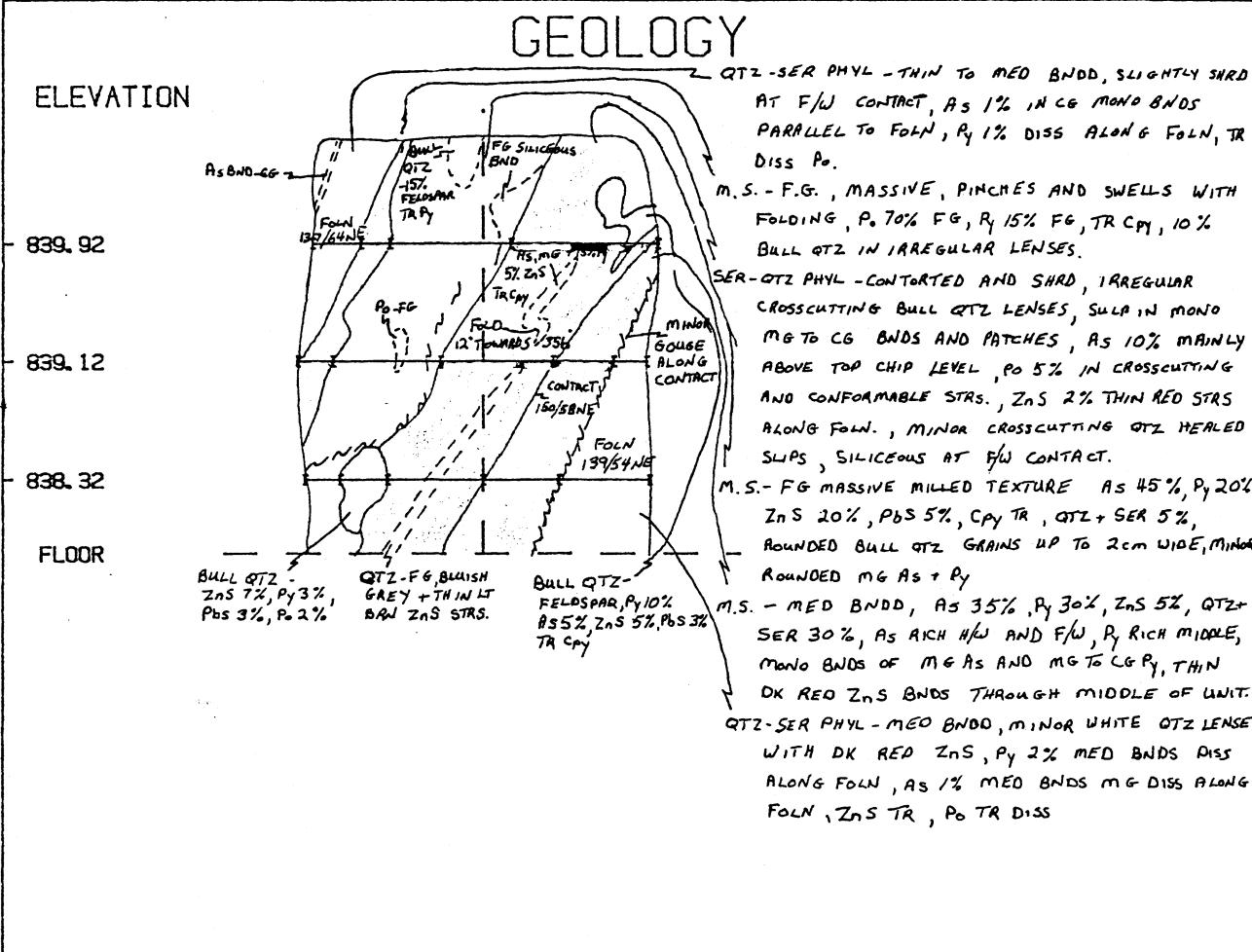
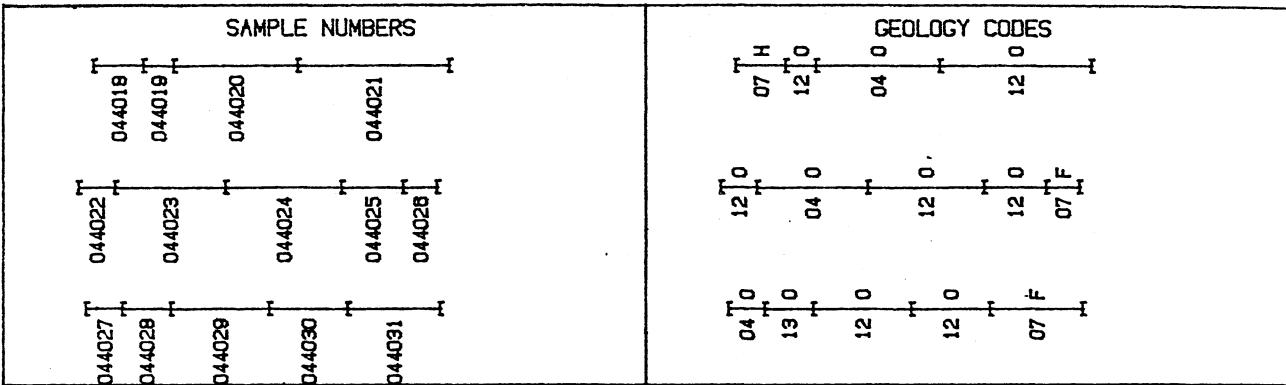


SELCOO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 166

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SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44018	.33	.03	.02	.448	.4	.3
44019	.20	1.05	3.15	3.960	18.1	2.5
44020	.82	.08	.50	.811	2.7	.7
44021	1.00	10.70	17.90	4.390	168.3	10.6
44022	.24	1.34	5.06	1.920	31.8	3.9
44023	.73	.08	.12	1.090	1.2	.9
44024	.77	13.70	18.30	2.580	326.2	9.7
44025	.41	6.31	10.70	8.470	96.9	12.1
44026	.22	.26	.64	5.600	9.7	3.3

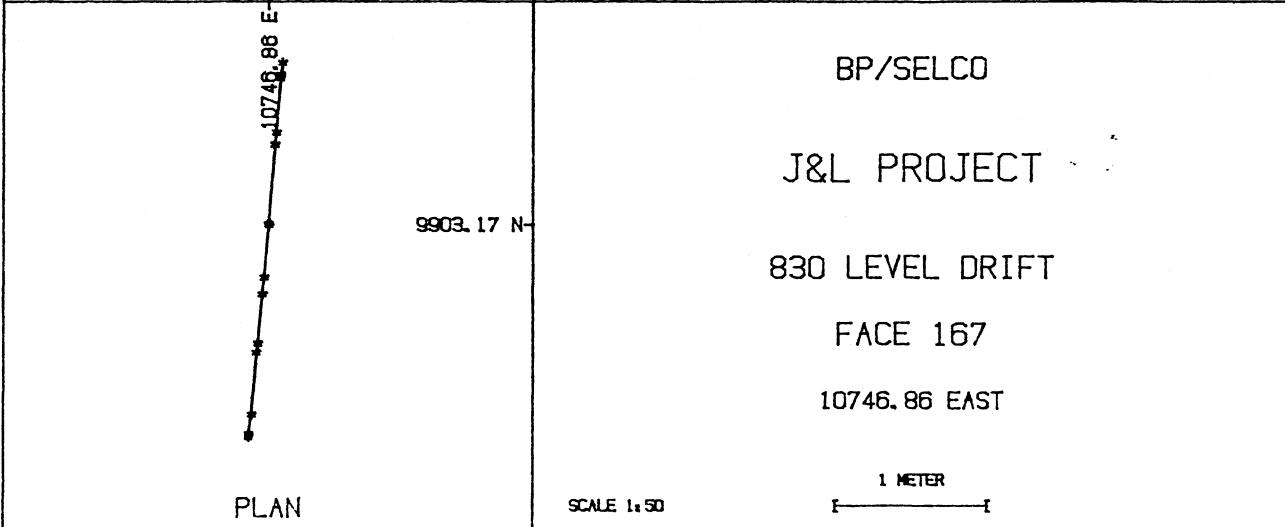
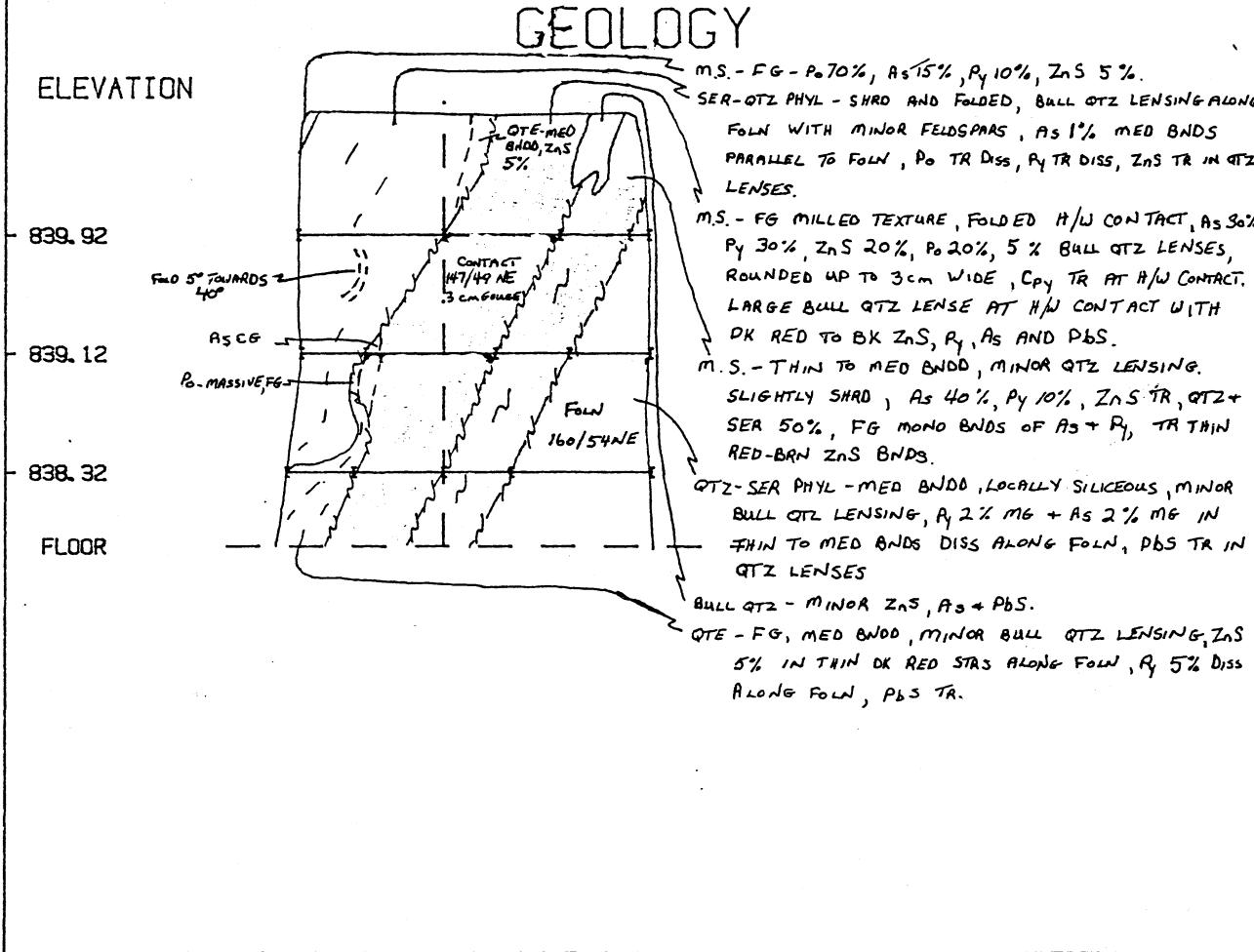
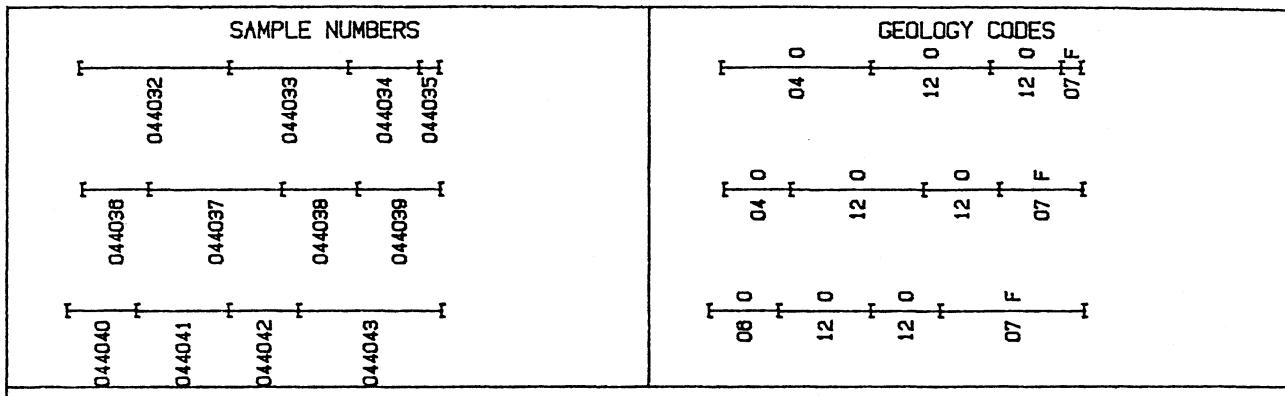


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 167

* * * * *

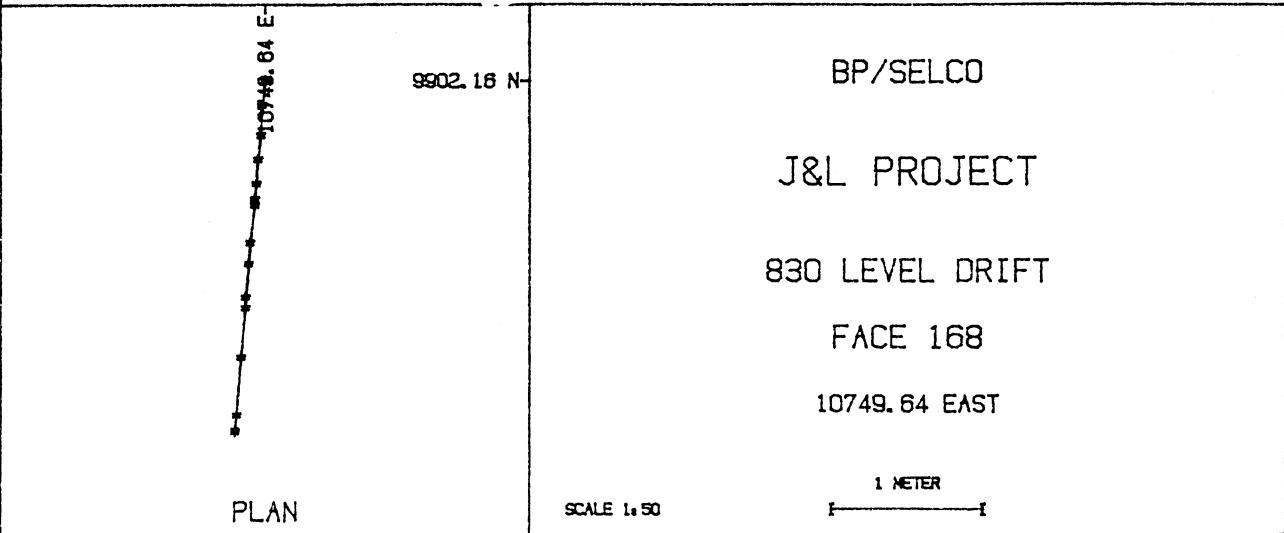
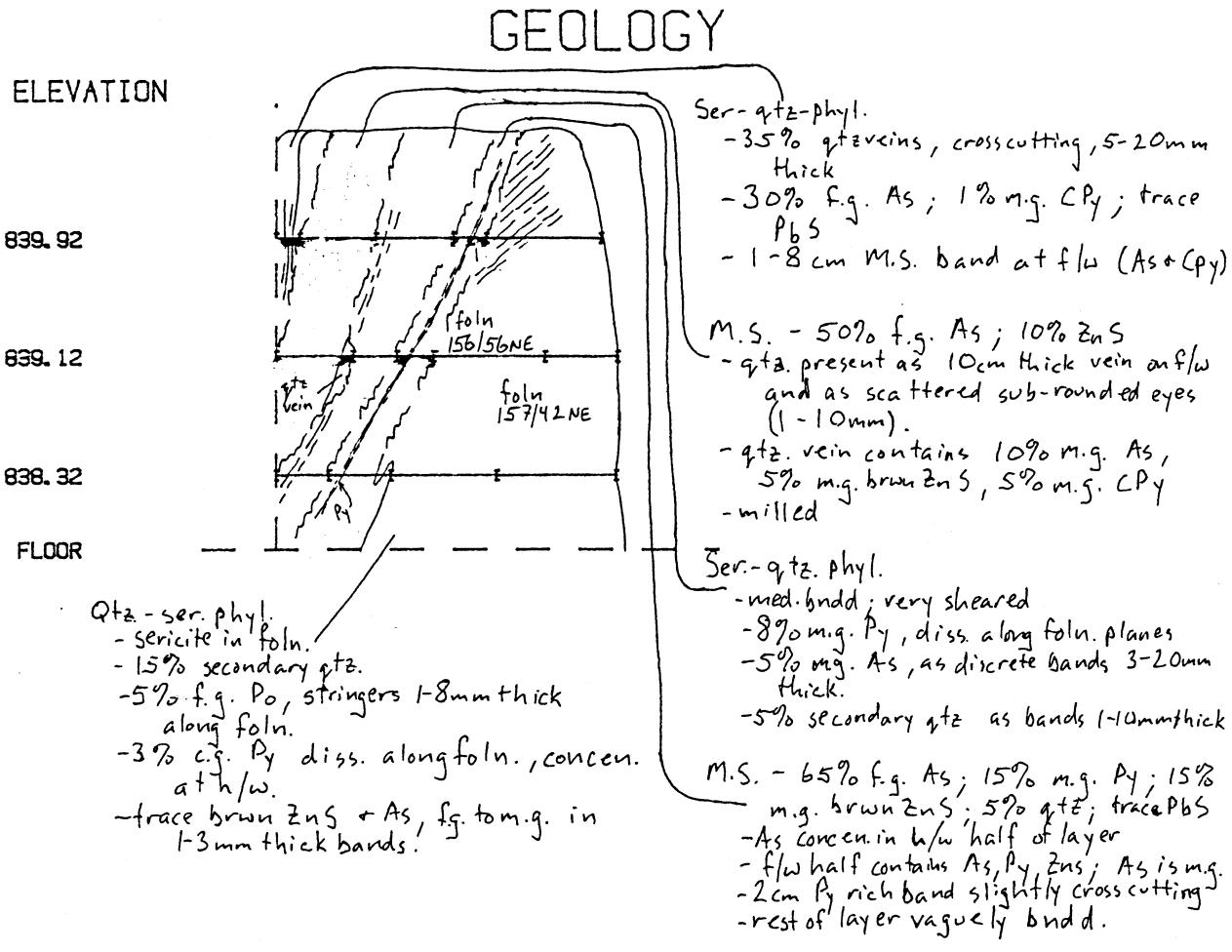
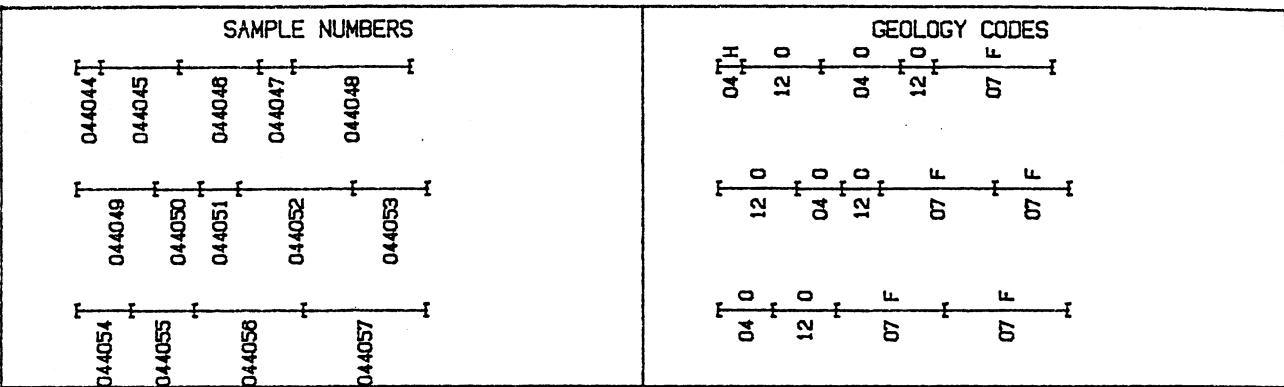
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44032	.99	.10	.22	1.810	5.4	.7
44033	.79	13.20	16.30	2.430	213.6	7.2
44034	.47	.38	.15	15.200	11.7	2.0
44035	.13	.12	.18	.414	4.8	.1
44036	.44	.30	.19	.108	15.8	.1
44037	.88	10.80	13.90	8.490	179.2	19.6
44038	.50	.25	.15	10.800	7.7	1.8
44039	.55	.03	.03	.173	3.9	2.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 168

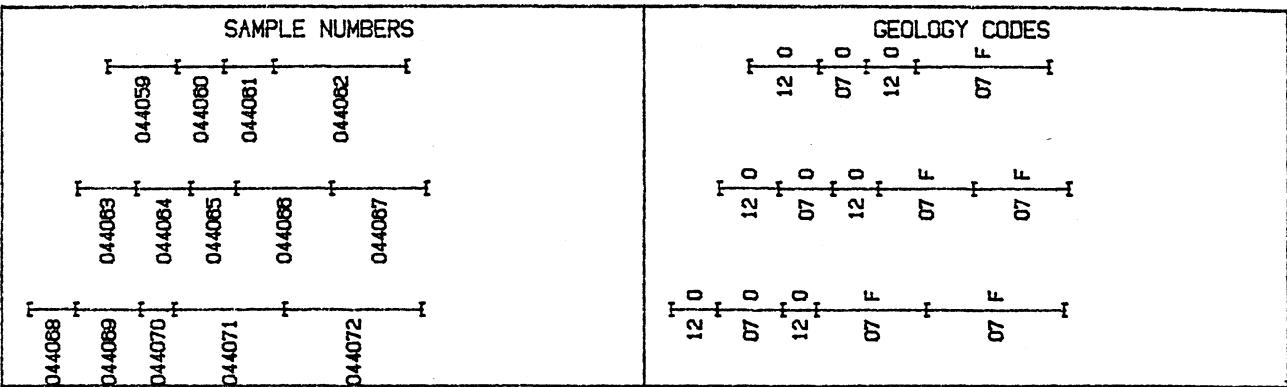
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44044	.16	.15	.64	.041	21.8	4.9
44045	.52	8.67	17.90	2.910	129.7	11.5
44046	.53	.47	.63	7.620	7.2	1.7
44047	.22	2.16	1.74	17.400	66.0	10.1
44048	.78	.06	.04	.435	2.4	.3
44049	.52	9.97	19.90	4.270	163.3	11.5
44050	.30	.23	.34	2.380	3.8	1.0
44051	.25	1.07	1.54	18.700	29.6	6.7
44052	.76	.03	.02	.586	2.9	.5
44053	.49	.01	.01	.184	2.5	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 169

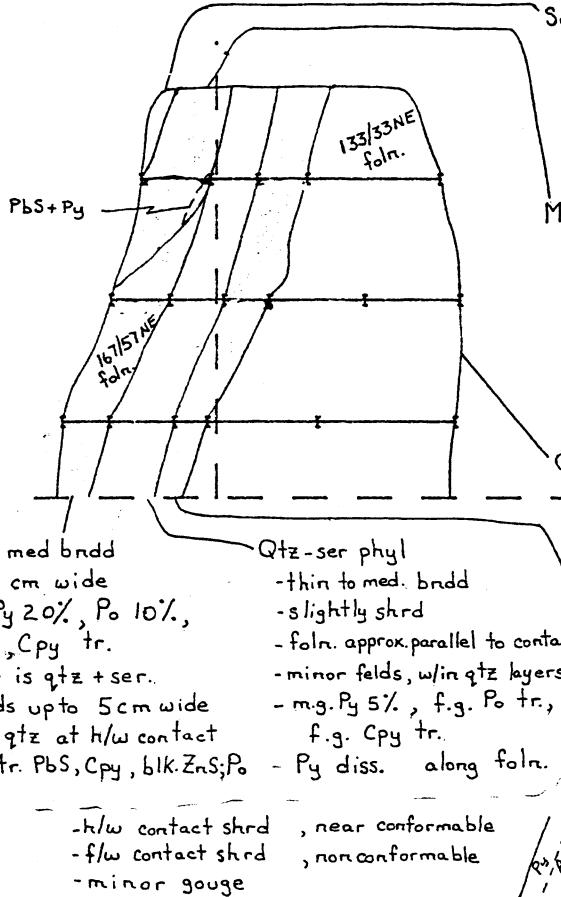
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44059	.46	13.20	17.90	6.500	206.4	7.5
44060	.31	.20	.24	.164	3.9	.2
44061	.33	8.97	7.16	2.720	203.3	7.9
44062	.88	.12	.09	.542	7.2	.3
44063	.39	.33	.61	6.020	6.4	1.9
44064	.36	.04	.03	.236	1.7	.3
44065	.30	5.21	.96	7.410	130.0	8.5
44066	.63	.54	.92	1.160	16.0	.4
44067	.63	.03	.03	1.220	32.7	.2



GEOLOGY

ELEVATION

840.03



FLOOR

M.S. - m.g., med bndd
- bands 1-5 cm wide
- As 30%, Py 20%, Po 10%,
ZnS 2%, CPy tr.
- remainder is Qtz + ser.
- sulf bands upto 5cm wide
- secondary Qtz at h/w contact
contains tr. PbS, CPy, blk ZnS; Po

Qtz-ser phyl
-thin to med. bndd
-slightly shrd
-foln. approx. parallel to contacts
-minor felds, w/in Qtz layers
-m.g. Py 5%, f.g. Po tr.,
f.g. CPy tr.
-Py diss. along foln.

Ser-qtz phyl thin bndd.
-slightly shrd. at sulf contact, x-cutting Qtz veins
1 cm thick, ~10cm long at contact
-Sulf in bands parallel to foln. - As (c.g., 15%)
Py (c.g., 5%), ZnS (c.g., red-brn, 1%).
- minor secondary Qtz

M.S. - f.g.
- As 40%, Py 30%, ZnS (brn) 20%,
PbS 2%.
- 8% white Qtz eyes, 1mm-2cm.
- minor re-xtal on margin. Patchy Py
+ PbS along margin
- grain size coarsens towards margin

Qtz-ser. phyl - med bndd
- locally sil.
- 5% Py, c.g. } in bands 1-20mm thick
- 2% As, m.g. } parallel to foln.
- 2% Po, m.g.
- 1% ZnS, f.g., red-brn, thin str.
- tr. CPy, f.g.

M.S. - vague to med bndd
- minor secondary Qtz patches containing
blk ZnS (c.g.), PBS (m.g.), CPy (f.g.), Po (m.g.)
- Qtz is at or near h/w contact
- As 50% (f.g.), ZnS 15% (f.g.), PBS 5%
(m.g.), Py 30% (f.m.g.)
- Py predom margins, As predom. core

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 169

10752.49 EAST

PLAN

9900.27 N

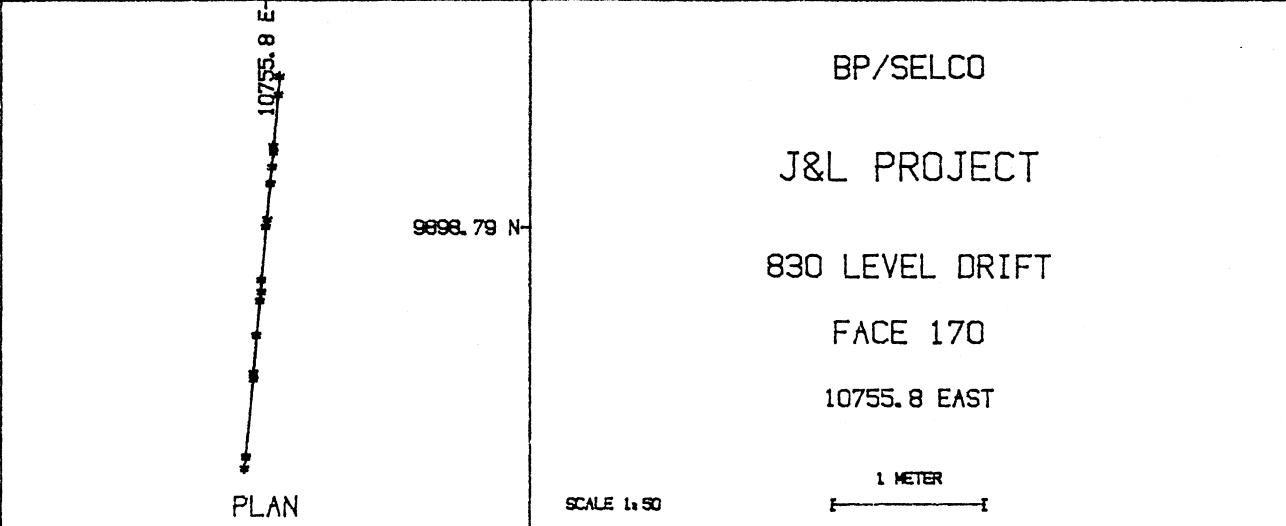
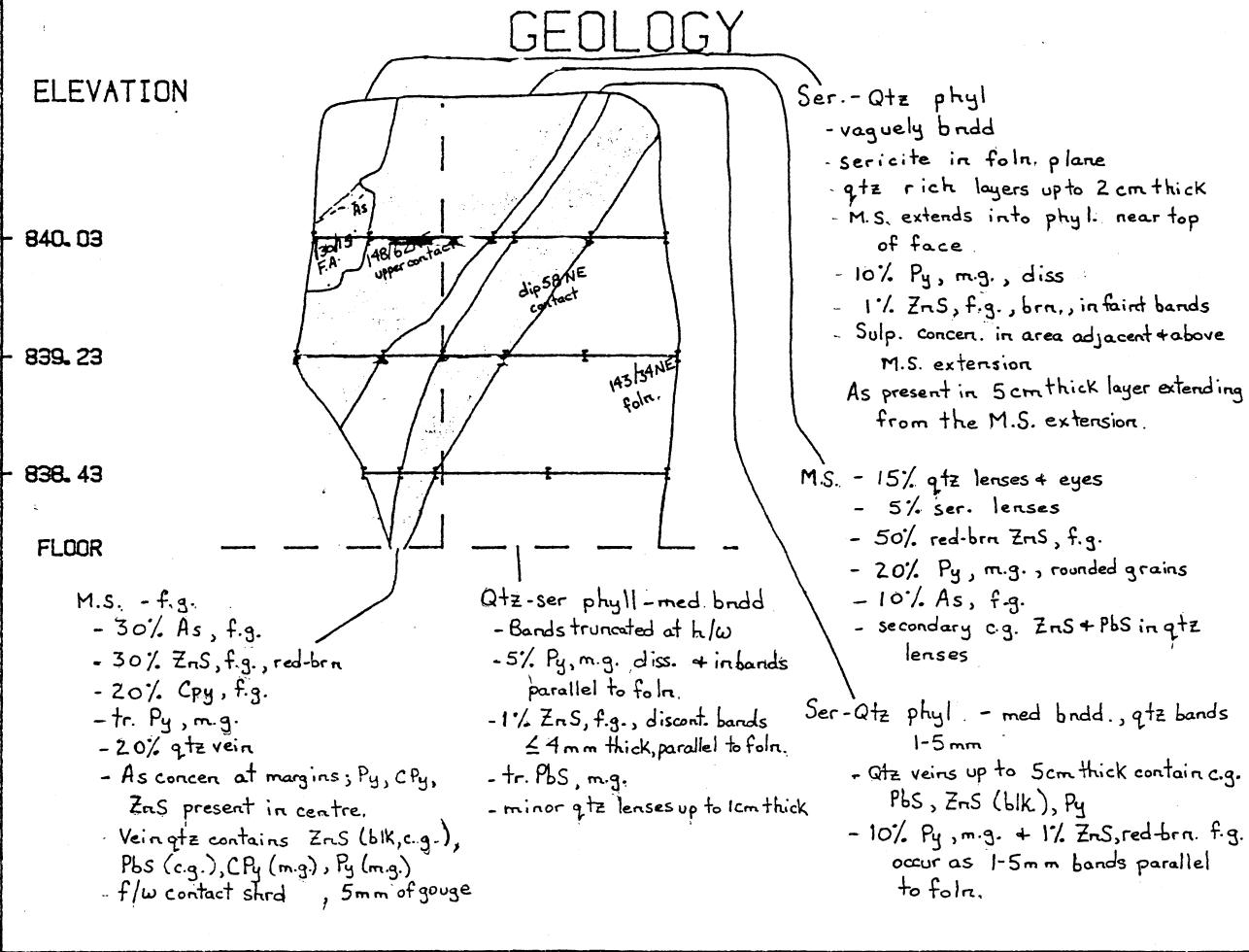
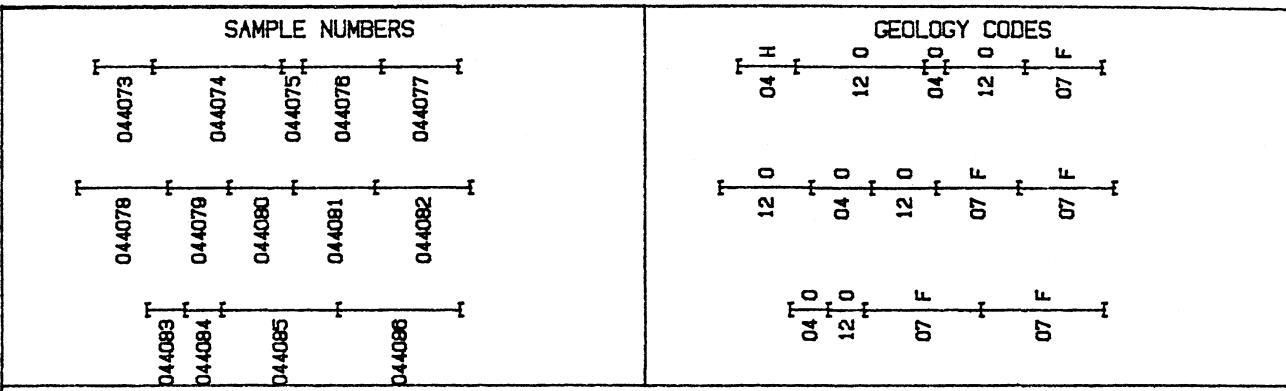
SCALE 1:50

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 170

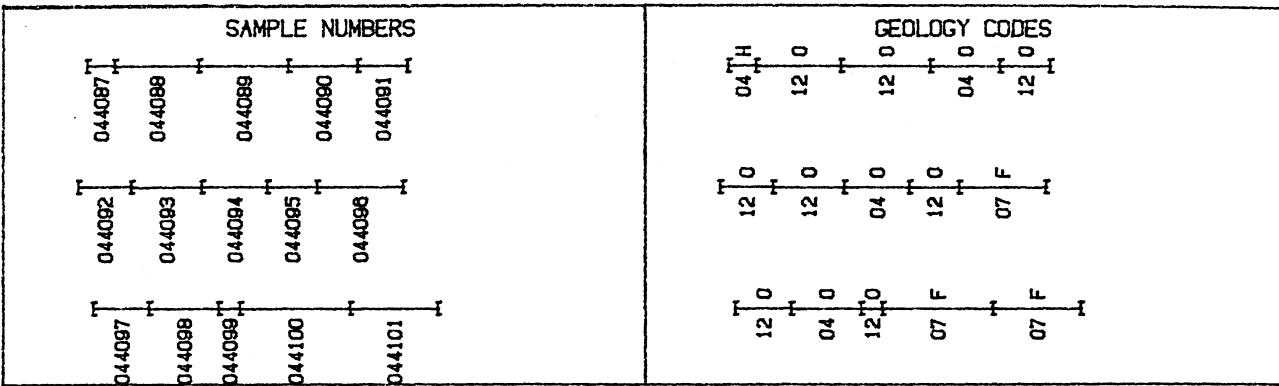
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44073	.38	.23	.15	3.530	6.1	3.5
44074	.85	12.20	18.60	2.170	232.8	7.9
44075	.19	.74	1.04	2.530	19.5	1.8
44076	.52	7.64	8.22	9.330	184.4	14.4
44077	.51	.13	.20	.192	4.6	.2
44078	.60	11.50	19.20	1.930	208.1	7.2
44079	.40	1.93	2.12	2.710	50.5	1.0
44080	.43	2.19	7.05	12.500	54.0	14.5
44081	.54	.23	.58	.332	7.8	.4
44082	.63	.30	.55	.119	11.4	.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 171

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44087	.18	1.57	2.61	1.450	28.4	1.1
44088	.56	9.65	11.10	3.740	154.3	11.7
44089	.59	8.37	8.86	3.340	160.0	4.5
44090	.46	2.84	3.71	.457	63.4	.3
44091	.33	6.24	5.40	18.100	159.6	20.7
44092	.35	10.70	15.80	3.210	193.3	10.3
44093	.47	12.40	15.00	2.440	233.2	5.4
44094	.43	7.17	5.74	.595	138.8	1.1
44095	.33	7.17	12.70	11.900	154.5	19.6
44096	.57	.32	.32	.418	3.6	.5



GEOLOGY

ELEVATION

840.03

839.23

838.43

FLOOR

Ser.-qtz. phyll.
- vaguely banded, quartz.
bands 1-5 mm.
- ser. in foln. plane
- sulf 5%, m.g.;
red-brown ZnS, PbS, Py
- sulf in x-cutting stringers
with quartz + as discord.
bands parallel to foln.

M.S. - f.g., milled
- quartz eyes ≤ 1 cm, lenses
up to 3 cm thick (5%)
- 50% ZnS, red-brown, f.g.
- 10% Py, rounded grains ≤ 3 mm
- 35% As, f.g.
- tr. PbS, m.g.
- at h/w contact blk ZnS (c.g.)
in f.g. PbS, Py, CPy
- f/w contact fairly sharp, ZnS
decreases over ~5 cm across
contact.

Ser.-qtz. phyll. - vague banding, 1-10 mm
- ser. in foln. plane
- 25% c.g. quartz lenses, contain c.g. ZnS (blk),
PbS, CPy, Py
- foln. wraps around quartz lenses.
- 10% Py, 1% ZnS (red-brown), 1% As, f.g. in disc.
layers parallel to foln. & diss.

M.S. - f.g.
- 35% Py, m.g., rounded grains ≤ 2 mm
- 30% As, f.g.
- 20% ZnS, red-brown, f.g. + smal. amt. c.g.
- 10% CPy, f.g.
- 5% quartz, small eyes 1-10 mm
- f/w contact slip surface, 3 mm gouge

Qtz.-ser. phyll - vague bndd.
- 5% Py, m.g., in bands 1-3 mm
- 1% ZnS, red-brown in bands w/ Py
M.S. - f.g. - ser. in foln. plane
- quartz eyes 2-10 mm, less
common than upper half
- patchy quartz at f/w contact (5%)
- 40% As, f.g.
- 35% Py, most f.g., some mg.
rounded grains.
- 15% CPy, f.g.
- 15% ZnS, red-brown, f.g.
- tr. PbS

10758.64 E

9897.53 N

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 171

10758.64 EAST

PLAN

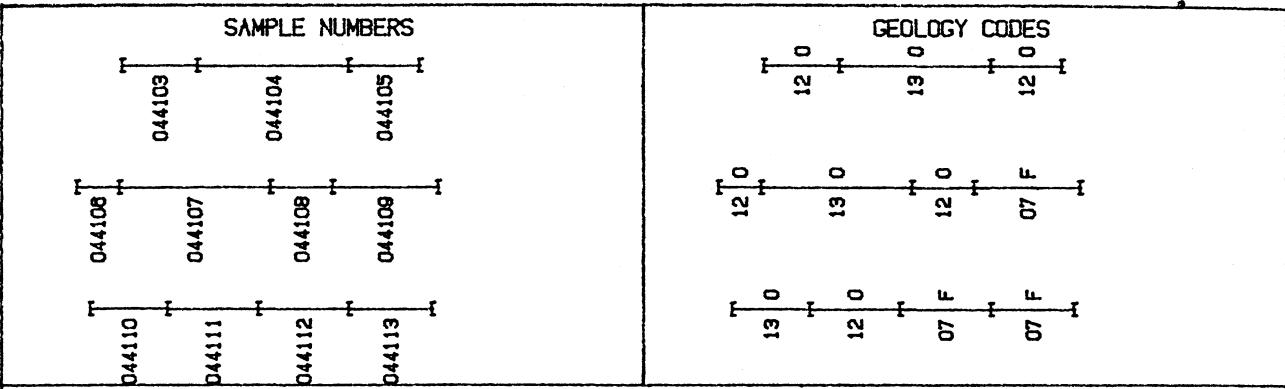
SCALE 1:50

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

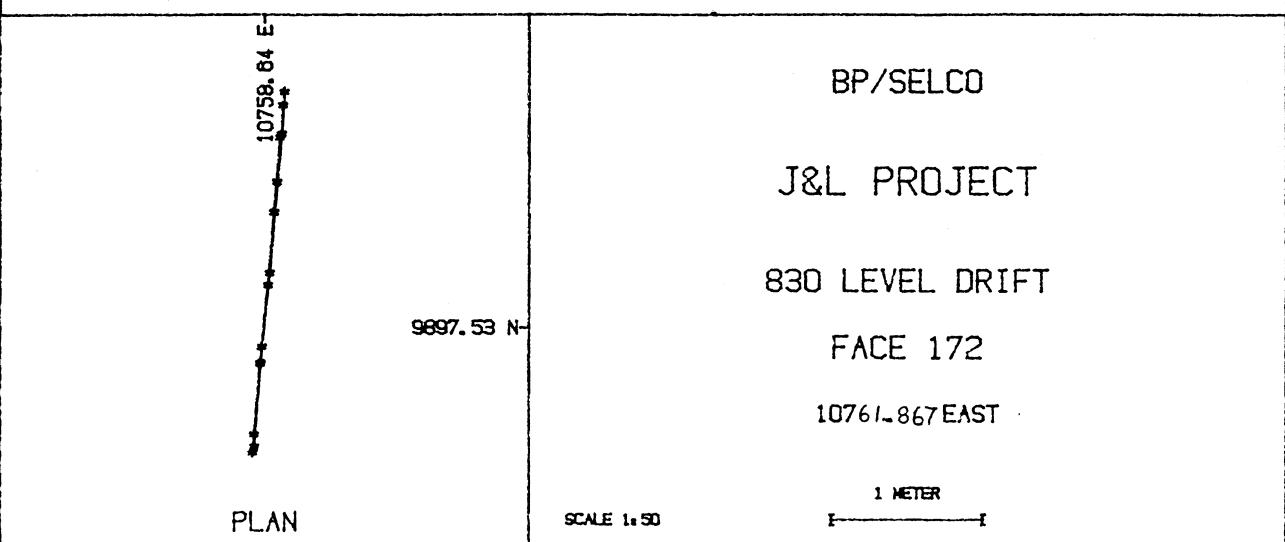
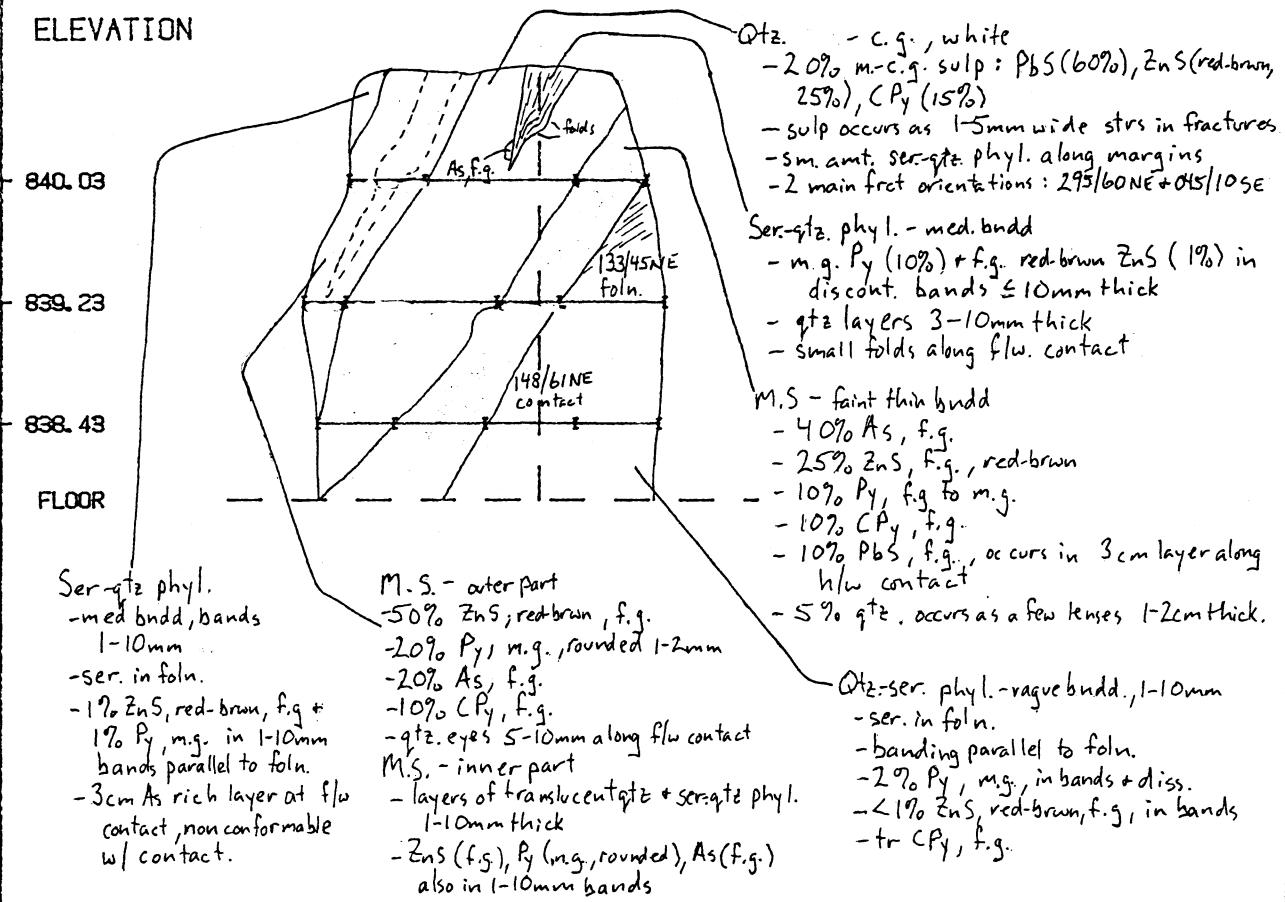
FACE# 172

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44103	.50	7.91	16.30	3.030	132.9	8.3
44104	1.00	9.11	8.67	.051	170.1	1.1
44105	.47	18.70	11.10	7.670	409.6	7.9
44106	.28	6.73	16.30	2.990	104.8	4.9
44107	1.00	7.99	1.64	.177	159.5	3.7
44108	.41	5.92	14.60	8.590	107.7	9.5
44109	.70	.40	.51	.284	12.7	.3



GEOLOGY

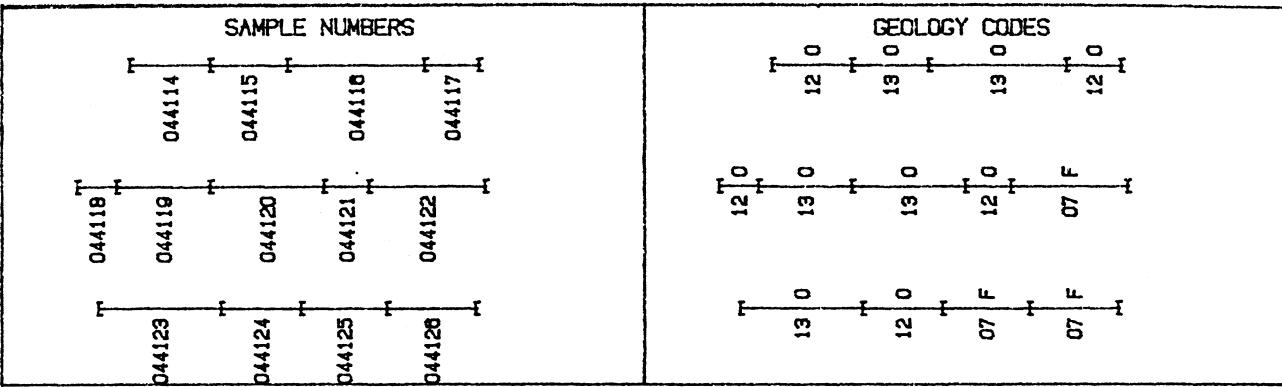
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 173

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44114	.53	10.60	16.00	1.840	142.6	6.2
44115	.51	9.09	8.52	.111	176.9	.7
44116	.91	8.21	10.30	1.520	186.9	3.0
44117	.36	12.80	3.78	8.120	281.7	11.7
44118	.26	7.30	14.10	5.050	213.2	8.9
44119	.62	3.36	3.72	.132	78.9	1.3
44120	.75	11.10	1.70	.040	254.1	1.6
44121	.30	3.96	16.80	9.700	98.8	10.2
44122	.76	.28	.38	.288	9.4	.2



GEOLOGY

ELEVATION

840.03

839.23

838.43

FLOOR

Ser.-qtz. phyl.

- med bndd.
- 2% ZnS, f.g., red-brown, in bands 1-5mm thick
- <1% Py, m.g., diss.
- some As starting 20cm above flw contact.

m.s.

- faint med. bndd.
- 50% ZnS, f.g., red-brown
- 20% Py, m.g., rounded, concen. 20cm above flw
- 20% As, f.g.
- 5% CPy, f.g.
- 5% qtz, wh. eyes 1-10mm + translucent in one fracture

Qtz. - wh. c.g.

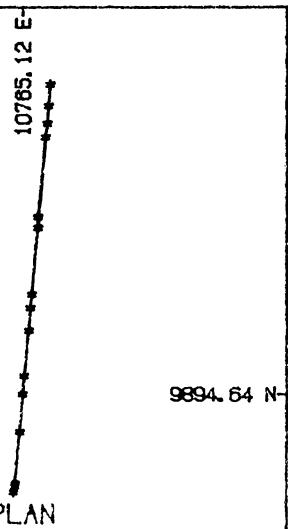
- 15% sulp: PbS (35%), CPy (15%), ZnS (red-brown, 50%)
- sulp. predom c.g., some m.g.
- sulp. as strs filling fractures
- most sulp w/in 20cm of margin
- core at top is essentially barren.
- some lenses of the qtz. in m.s. along both margins.

M.S. - faint med. bndd.

- 35% Ats, f.g.
- 35% ZnS, red-brown, f.g.
- 20% Py, m.g.
- 5% PbS, f.-m.g.
- tr. Po, f.g.
- Py concen. in upper half of layer
- lenses of qtz. w/ c.g. sulp as in qtz above

Qtz.-ser. phyl.

- ser. in foln.
- 5% qtz lenses upto 2cm thick, 10cm long
- 2% ZnS, f.g., red-brown, diss. + faint thin bndd.
- tr. PbS, m.g., diss.
- tr. Po, f.g., in foln.
- foln. slightly discordant w/ sulp.
- ~1cm gauge, h/w contact



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 173

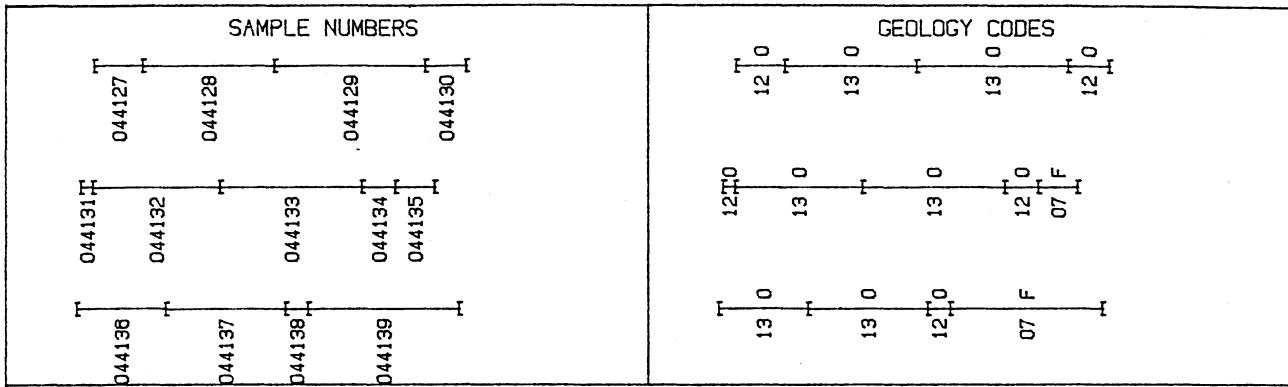
10765.12 EAST

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 174

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44127	.32	11.70	18.60	2.390	181.4	7.9
44128	.87	7.36	.61	.035	165.4	.5
44129	1.00	2.52	2.92	.347	59.9	.4
44130	.27	2.37	3.05	8.450	47.3	5.5
44131	.08	11.70	17.60	2.590	185.6	6.4
44132	.84	.76	.77	.039	17.8	.7
44133	.94	2.33	3.36	2.740	60.6	2.5
44134	.22	.30	3.15	5.570	10.5	2.5
44135	.26	.05	.08	.682	2.4	.3



GEOLOGY

ELEVATION

840.14

839.34

838.54

FLOOR

Ser.-qtz phyl.

- med. bndd
- 3% As, f.g.; 2% Py, m.g.; <1% ZnS, red-brown, f.g.
- lower 8cm: ZnS, layers 1-6mm; Py, f.-m.g., layers \leq 1cm.
- next 20cm: tr. Py, diss.
- \geq 28cm above f/w: 5% As, f.g., diss.

M.S. - f.g.

- 60% ZnS, f.g., red-brown
- 30% As, f.g.
- 5% Py, m.g., rounded
- 5% qtz eyes 1-30mm
- wh. qtz w/ c.g. sulp along f/w, lenses \leq 15cm long

Qtz. - c.g., wh.

- 5% c.g. sulp: 40% PbS, 40% red-brown ZnS, 20% CPy, <1% Po
- sulp. concen. w/in 10cm of margin, some in centre.
- sulp. in fractures, 1-5mm thick
- a few thin, disc. layers ser.-qtz. phyl. near f/w, $\frac{1}{2}$ -1cm thick
- 3-5cm thick layer ser.-qtz. phyl. along f/w.
- M.S. - med. bndd, layers 1-10mm
- 40% Py, f.-m.g.
- 30% As, f.g.
- 25% ZnS, f.g., red-brown
- 5% wh. qtz w/ c.g. sulp in upper $\frac{1}{2}$, wh. qtz eyes lower $\frac{1}{2}$ + thin layers 5-5mm
- Py. concen. in upper $\frac{2}{3}$.

Qtz. ser. phyl. - med. bndd.

- ser. in foln.
- qtz. lenses 1-8mm thick
- <1% Py, m.g., diss.
- <1% ZnS, f.g., red-brown, bands 1-2mm thick
- 1 cm gouge on h/w
- foln. \sim parallel to contact w/m.s.

10768.49 E-

9894.33 N-

PLAN

SCALE 1:50

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 174

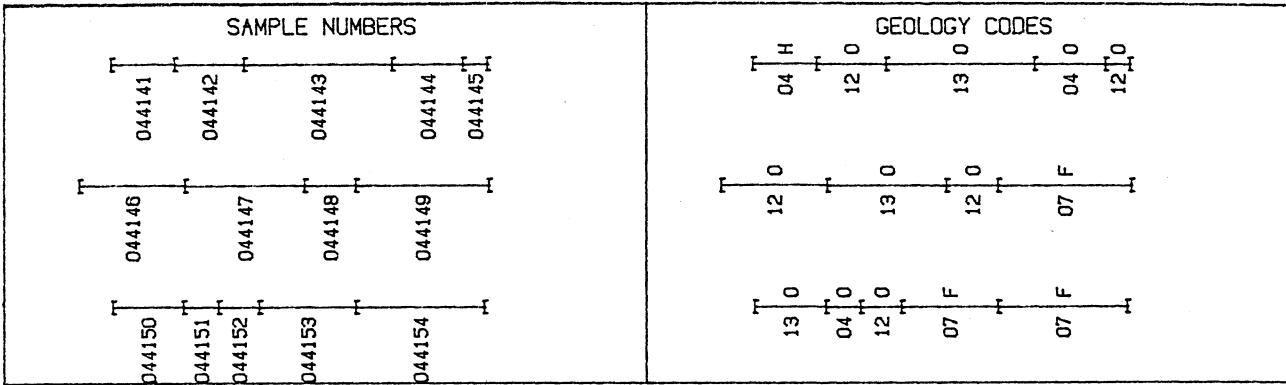
10768.49 EAST

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

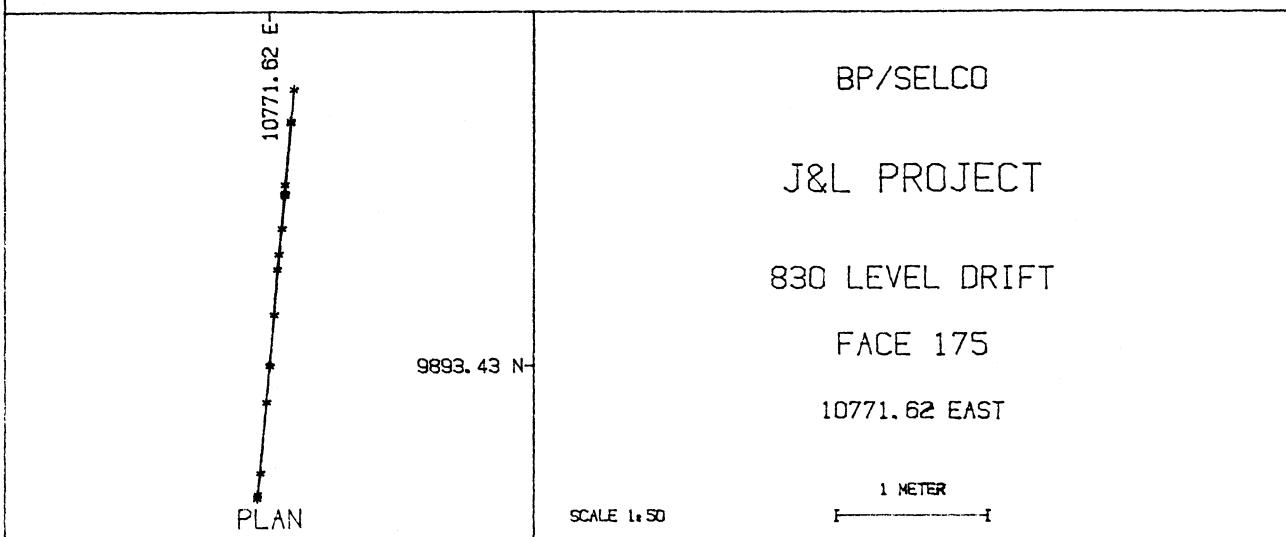
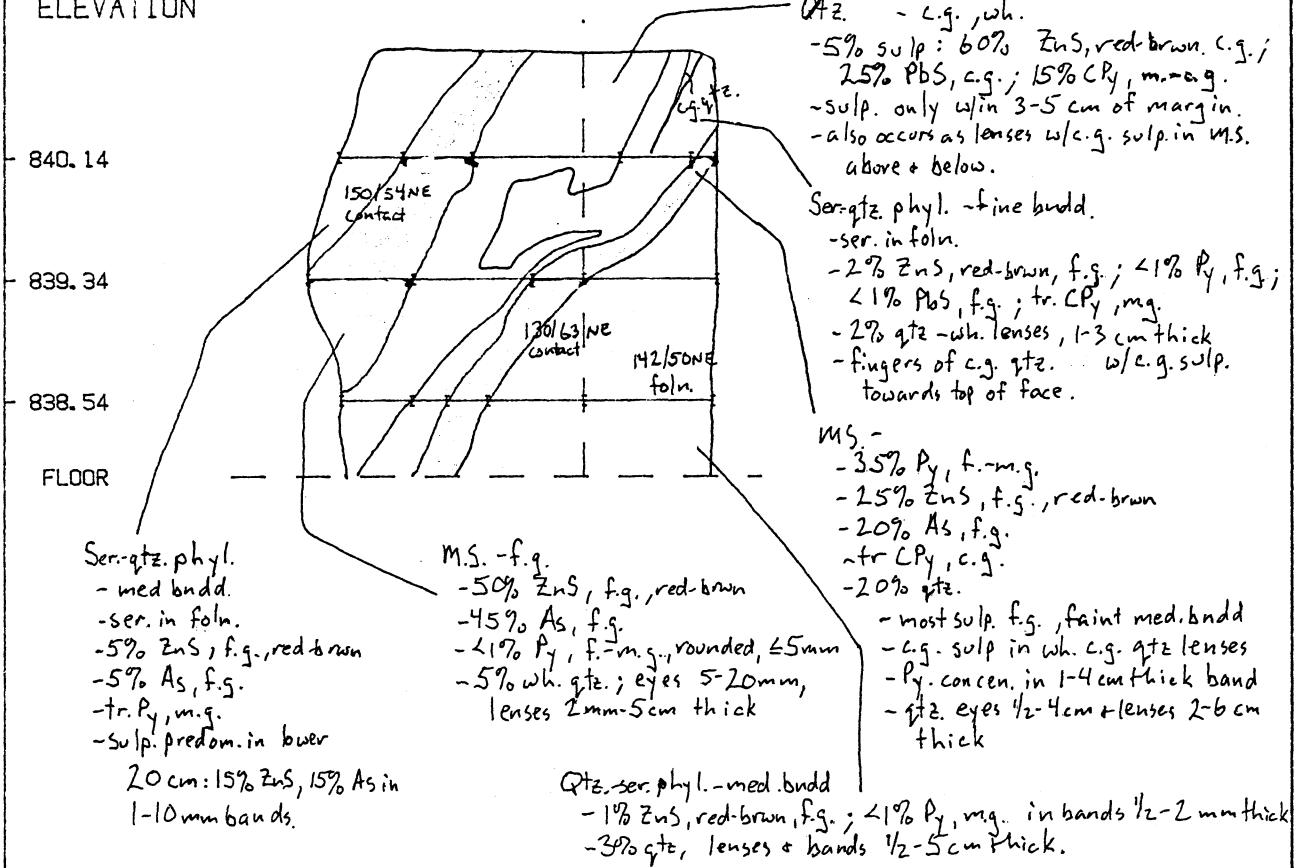
FACE# 175

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44141	.42	1.75	7.05	.174	25.1	.9
44142	.46	13.40	17.90	2.100	200.6	8.5
44143	.98	.56	1.08	.029	14.8	.3
44144	.47	1.34	4.19	1.300	25.7	1.0
44145	.16	4.20	10.70	4.140	91.8	4.9
44146	.70	12.50	19.90	2.250	186.4	7.6
44147	.79	3.05	7.68	.734	63.6	.8
44148	.34	10.80	14.10	6.370	238.0	8.8
44149	.88	.35	.58	1.050	10.7	.7



GEOLOGY

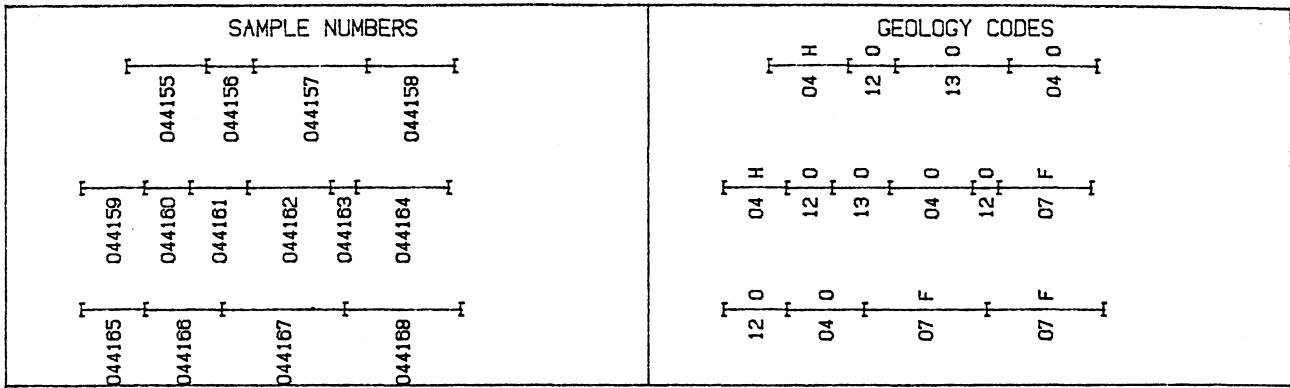
ELEVATION



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 176

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44155	.51	.21	.46	1.140	7.8	.4
44156	.31	11.70	18.30	2.040	153.6	7.5
44157	.75	3.80	5.74	.257	76.0	2.8
44158	.58	6.36	3.10	.788	135.9	.5
44159	.42	.25	.76	.606	7.5	.7
44160	.30	12.06	18.90	2.130	186.7	7.3
44161	.38	3.62	5.92	.926	66.4	.8
44162	.55	2.78	7.16	.453	58.6	.4
44163	.17	3.32	5.92	4.770	80.8	3.5
44164	.61	.10	.13	.224	6.0	.2



GEOLOGY

ELEVATION

840.14

839.34

838.54

FLOOR

Ser-gte. phyl.

- med. bndd.
- 1% ZnS, f.g., red-brown,
- discon. bands 1-2mm thick
- <1% Py, f.-m.g., diss.
- banding parallel to foln.,
foln. ~ parallel to f/w
contact.

M.S.-f.g.

- 60% ZnS, f.g., red-brown
- 40% As, f.g.
- <1% Py, f.-m.g., rounded
- 2% qtz. eyes, wh., 1-20mm
- f/w contact w/ gte.
irreg.

Qtz. -wh., c.g.

- sulp. 10%: 40% ZnS, c.g., red-brown;
- 40% PbS, m.-c.g.; 20% Py, c.g.;
- tr. Po, c.g.
- sulp. in fractures, 1-5mm thick
- fractures predom. ~ parallel to f/w.

Ser-gte. phyl. -med. bndd. bands 1-5mm

- 5% As, f.-m.g., bands 2-10mm
- 3% ZnS, f.g., bands 1-2mm
- 2% Py, m.g., diss. & bands 1-5mm
- ser. in foln.

Ser-gte. phyl. / gte vein interfingered.

- both rock types as desc. above
- ser./gtz. ~ 50/50
- c.g. sulp in gte. mainly along margins
- a few lenses 1-2cm thick of f.g. M.S.,
same comp. as f/w M.S.

M.S. -f.g., med. bndd, bands 1-5mm

- 50% As, f.-m.g.
- 30% ZnS, f.g., red-brown
- 20% Py, f.-m.g.
- sulp. c.g. at contact w/ gte

Qtz.-ser. phyl -med. bndd

- 2% Py, f.-m.g.; <1% ZnS, f.g., red-brown
in bands 1-2mm thick
- gtz. lenses 1-5cm thick
- ser. in foln.

E

10774.81

9892.52 N

PLAN

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 176

10774.81 EAST

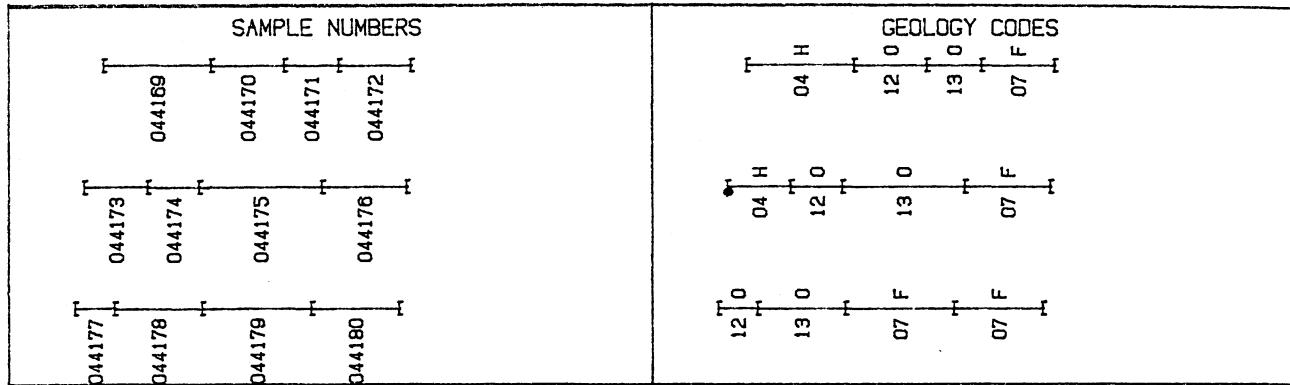
1 METER

SCALE 1:50

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 177

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44169	.71	.61	.68	1.330	15.3	.9
44170	.48	11.20	16.80	2.130	168.1	8.8
44171	.36	3.92	15.30	2.070	79.9	4.4
44172	.48	1.54	9.03	.810	34.1	.9
44173	.42	.16	.40	.687	.3	.6
44174	.34	11.40	19.50	2.090	174.9	6.8
44175	.81	5.24	9.20	.226	111.5	.9
44176	.56	.58	2.06	.226	16.1	.3



GEOLOGY

ELEVATION

840.14

839.34

838.54

FLOOR

Ser=qtz. phyl.
-med. bndl.
-scr in foln.
-2% Py, f-mg., diss +
bands 1-5 mm
-2% ZnS, f.g., red-brown,
bands 1-2 mm
-<1% As, f-mg., bands
1-5 mm
-5% qtz. wh., bands
5-10 mm

m.s. - f.g., faint med. bndl.
- 60% ZnS f.g., red-brown
- 30% As, f.g.
- 5% Py, f-mg., diss.
- 5% qtz. wh., eyes 5-20mm
+ lenses 1-5mm wide (sheared
eyes?)
- Py concen. top 2-5cm, also
several ZnS bands 2-5mm.
- bands parallel to contacts

Qtz. - wh., c.g.
- 10% sulf: 40% PbS, c.g.; 40% ZnS, red-brown,
c.g.; 20% CPy, c.g.; tr. Po, mg.
- sulf. in fractures, 1-5mm thick

Qtz-ser. phyl. - med. bndl.
- 5% Py, f-mg., diss + bands ≤ 1cm
- 1% ZnS, f.g., red-brown, bands 1-5mm
- tr. CPy, f.g., bands ≤ 5mm
- 5% wh. Qtz. bands + lenses 5-20mm
- ser. in foln.

Qtz. - wh., c.g.
- varies 1-10cm thick, parallel to foln.
- 35% sulf: 60% ZnS, red-brown, c.g.; 30% PbS,
m-c.g.; 10% CPy, c.g.
- f.g. wls along fw of qtz.
- faint med. bndl., bands 1-5mm
- 50% Py, f-mg.; 35% As, f.g.; 25%
ZnS, f.g.
- layer varies 0cm at top offace to
5cm at bottom.

10778.3 E

9891.52 N

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 177

10778.3 EAST

PLAN

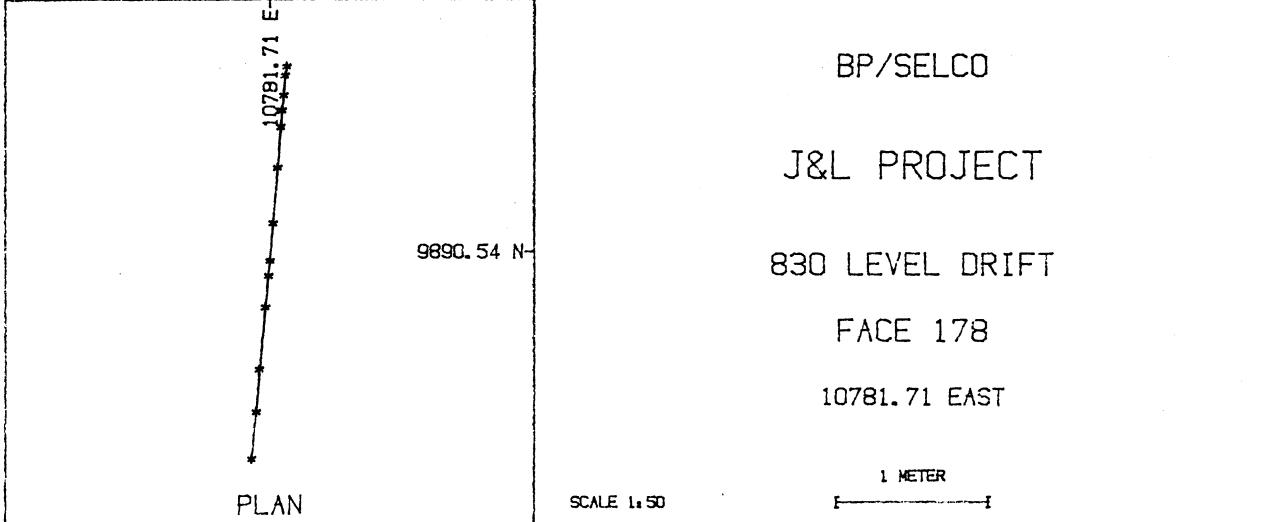
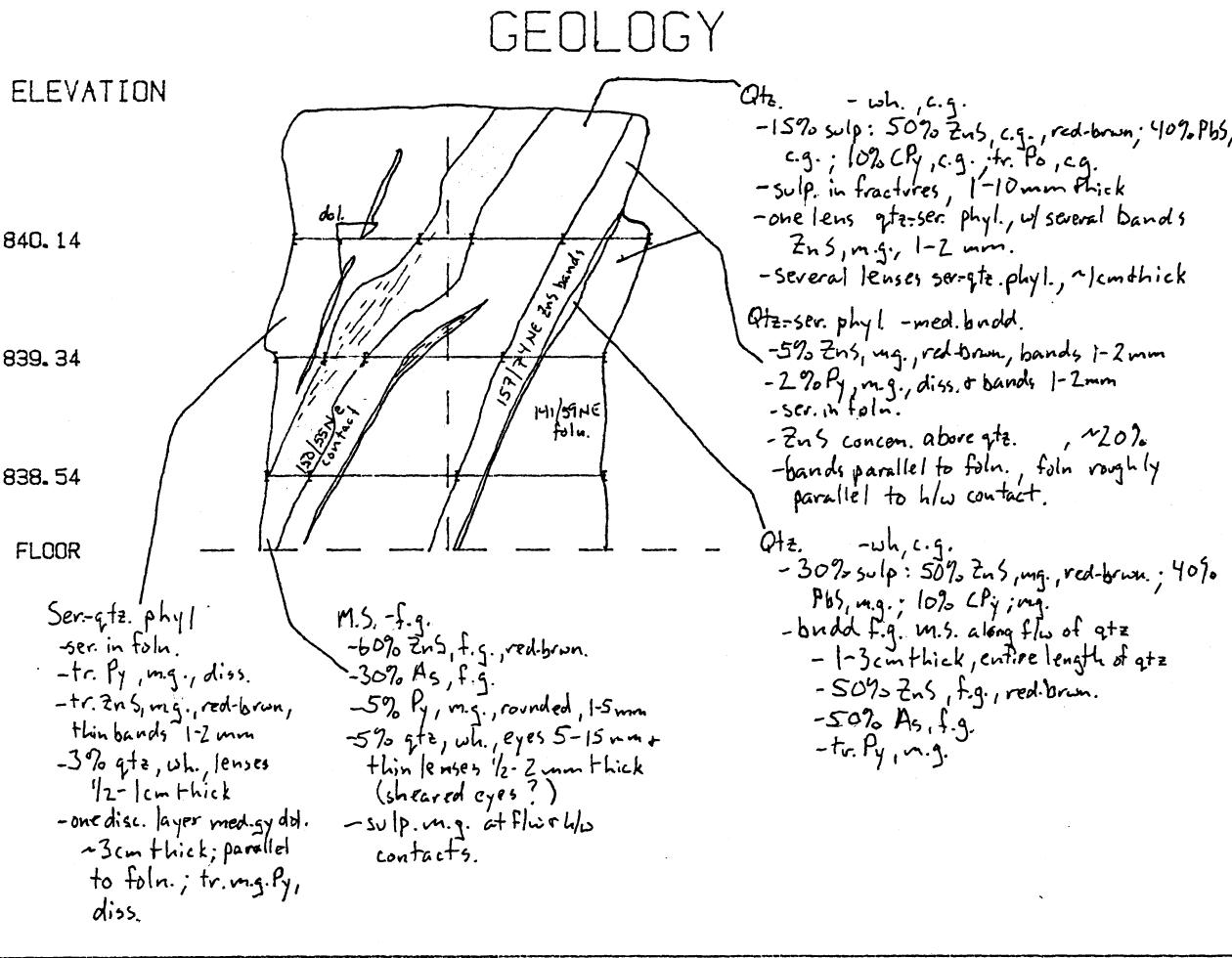
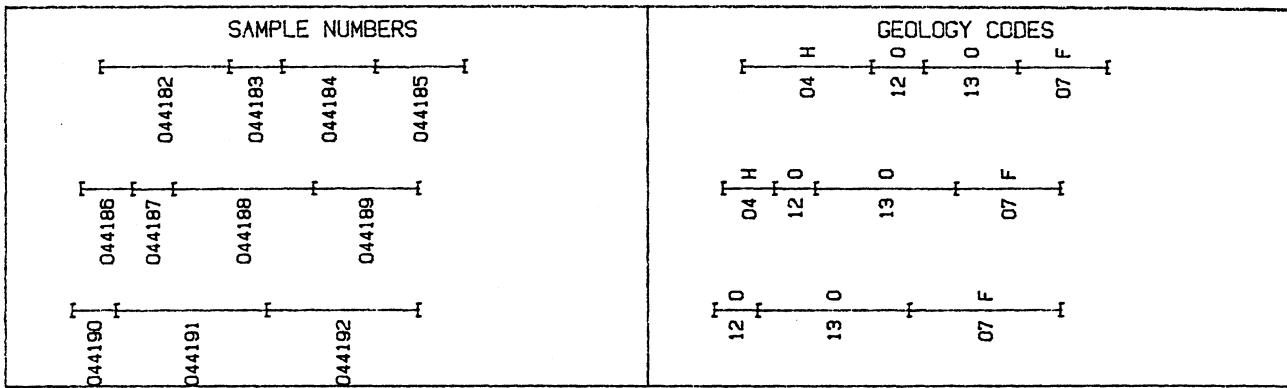
SCALE 1:50

1 METER

SELCO DIV. - BF RESOURCES J & L PROJECT, B.C.

FACE# 178

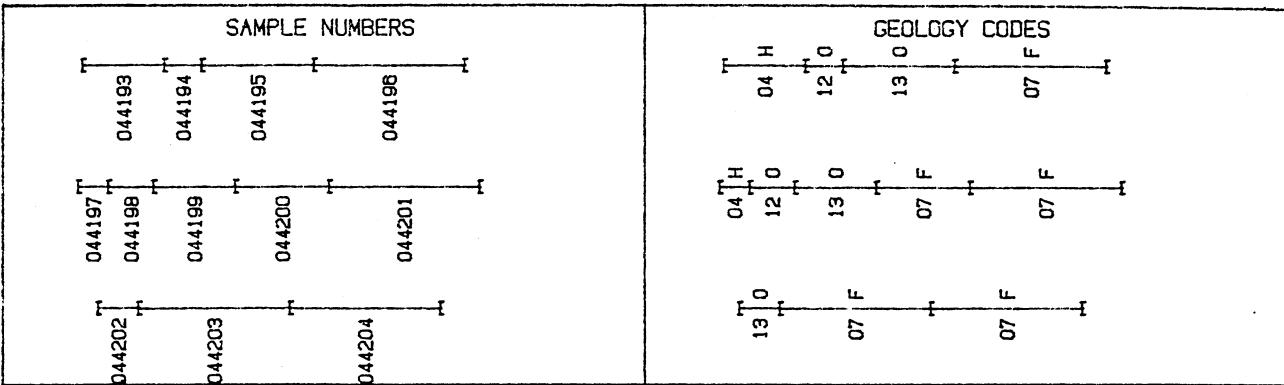
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44182	.85	3.62	11.30	1.390	93.4	2.5
44183	.35	4.90	13.30	1.960	85.4	3.0
44184	.62	11.40	5.74	.127	229.3	.4
44185	.59	3.92	6.10	1.620	79.5	1.4
44186	.34	.14	.15	.061	3.3	.1
44187	.27	11.00	17.90	2.390	177.1	6.6
44188	.93	4.00	9.39	.144	80.2	1.4
44189	.69	.86	4.11	1.510	29.3	.8



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

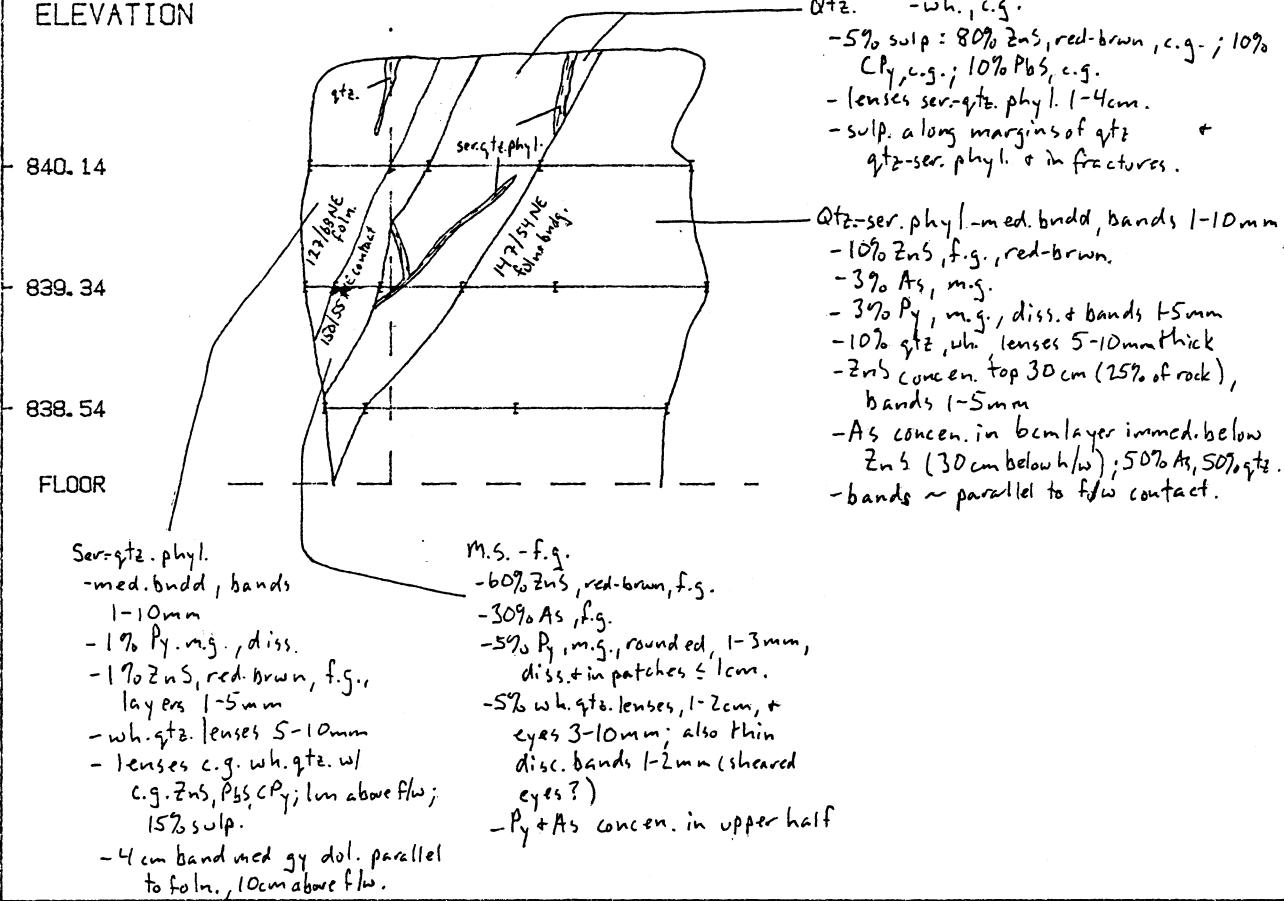
FACE# 179

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44193	.54	.01	2.17	.005	2.4	.4
44194	.25	8.96	19.90	2.810	47.0	7.7
44195	.74	.53	5.64	.186	.3	.3
44196	1.00	.42	2.06	7.940	8.6	.9
44197	.20	.02	.07	.075	3.8	1.0
44198	.30	3.08	20.20	3.870	69.6	9.9
44199	.54	.91	6.30	.141	1.4	.1
44200	.62	.17	11.50	.158	1.3	.1
44201	1.00	.01	.11	.028	1.3	1.4



GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 179

10784.79 EAST

1 METER

PLAN

SCALE 1:50

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 180

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44205	.61	.40	1.08	.230	.3	.4
44206	.30	5.24	16.60	1.840	98.6	3.5
44207	.69	.45	.82	5.850	7.5	1.3
44208	.80	.14	.06	2.520	.3	1.7
44209	.31	.09	.06	.229	.3	.2
44210	.24	5.81	21.70	1.910	104.9	5.5
44211	.70	.12	2.22	.192	.3	.3
44212	.68	.31	.13	5.030	10.6	3.1
44213	.78	.01	.04	.063	.3	.4

SAMPLE NUMBERS					GEOLOGY CODES						
044205	044206	044207	044208		04 H	12 0	F	07	07		
044209	044210	044211	044212	044213	04 H	12 0	F	07	F		
044214	044215	044216	044217	044218	12 0	F		07	F		
								07	F		

GEOLOGY

ELEVATION

840.14

839.34

838.54

FLOOR

Ser-gtz.phyl.
-slightly calcareous
-med.bndd, bands
parallel to foln.
-1% Pyrm-c.g., irreg.
bands 5-10mm
-tr. ZnS, f.g., red-brown,
thin disc. bands
1/2 - 1 mm.
-ser. in foln.

M.S. -f.g.
- 5% ZnS, f.g., red-brown.
- 40% As, f.g.
- 5% Py, m.g., diss.
- 5% wh. gte., eyes 2-20mm
- As+Py concen. lower 3 cm
- c.g. wh. gte. lenses along flw,
3-10mm thick
-- 2% c.g. sulp.: 40% CPy;
30% ZnS, 30% PbS
-- sulp. in fractures

Qtz. -wh., c.g.
- 2% c.g. sulp.: 6% PbS, 40% ZnS
(red-brown); sulp. in fractures

Qtz-ser. phyl. -med bndd
- 10% ZnS, red-brown, bands 1-5mm
- 1% Py, m-c.g., bands 5-10mm + diss.
- 1% As, f.g., diss.
- banding slightly discordant w/M.S.
- As+Py concen. along margin of
gtz.

Qtz-ser. phyl. -med.bndd.
-tr. ZnS, f.g., red-brown, thin bands 1/2 mm
- 1% Py, m.g.
- 1% As, f.g.
- Py+As concen. in top 3-15cm, sulp.
~30%
- wh. gte. lenses 5-15mm, parallel to
bands + foln.
- 1-3mm gouge along slit surface
- ser. in foln.

10787.77 E
PLAN

9888.8 N

BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 180

10787.77 EAST

SCALE 1:50

1 METER

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 181

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44219	.10	.35	.69	3.040	5.7	.5
44220	.27	8.98	17.90	2.360	146.9	8.7
44221	.26	3.80	13.50	1.270	75.5	3.3
44222	.53	.74	1.99	3.230	17.2	.6
44223	.88	.06	.12	.799	.7	.7
44224	.42	.14	.85	1.090	2.5	.9
44225	.81	.03	.05	.793	1.1	.9
44226	.78	.01	.04	.225	.3	.3

SAMPLE NUMBERS

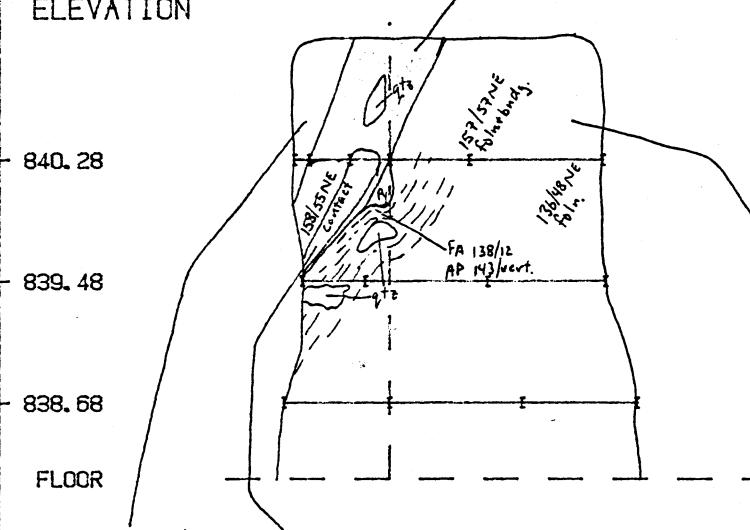
044219		
044220	044221	044222
		044223
044224		
	044225	044226
044227		
	044228	044229

GEOLOGY CODES

04	TH	12	0	H	F	F
07	F		07	F		
07	F		07	F		
07	F		07	F		

GEOLOGY

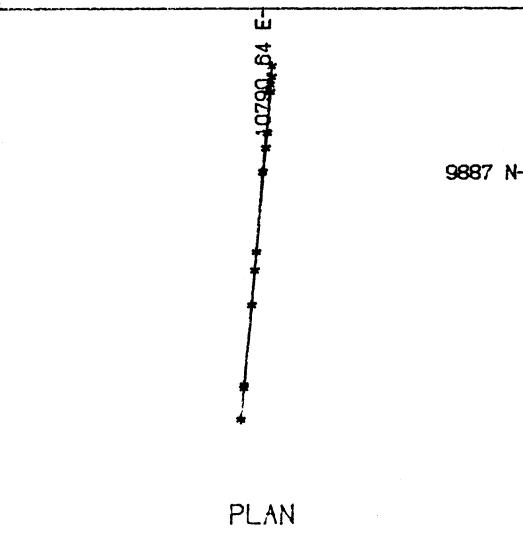
ELEVATION



Ser-gtz. phy.
- med. bndd, bands
1-10 mm
- tr ZnS; red-brown, f.g.,
bands 1-3mm
- tr. Py, f.g., diss.
- tr. As, f.g.
< ZnS predom. w/in
10 cm of flw
- slightly calcareous

Ser-gtz. phy.
- med. bndd, bands
1-20 cm
- tr ZnS, red-brown, f.g.,
bands 1-3mm
- tr. Py, f.m.g., disc. bands
1-5 mm + diss.
- banding contorted

M.S.-f.g.
- 50% ZnS, f.g., red-brown.
- 40% As, f.g.
- 5% Py, f.m.g., diss.
- 5% Qtz, wh., eyes 5-20mm
- Py concen. in bottom 4cm, patchy:
Qtz-ser. phy. - med. bndd, bands 1-5mm
- tr. Py, f.g., diss.
- tr. C Py, f.g., bands 2-5mm, disc.
- 5% wh. Qtz, lenses 5-20mm, parallel
to foln.
- Sulp. concen in upper 30 cm:
-- 10% Py, f.g.
-- 1% ZnS, bands 1-8mm
-- 1% As, bands 1-5mm
- top 40cm sheared
- two wh. Qtz lenses, c.g.
-- 2% sulp: 40% C Py, m.g.; 30% PbS, m.g.;
30% ZnS, f.g.
-- Sulp. in fractures



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

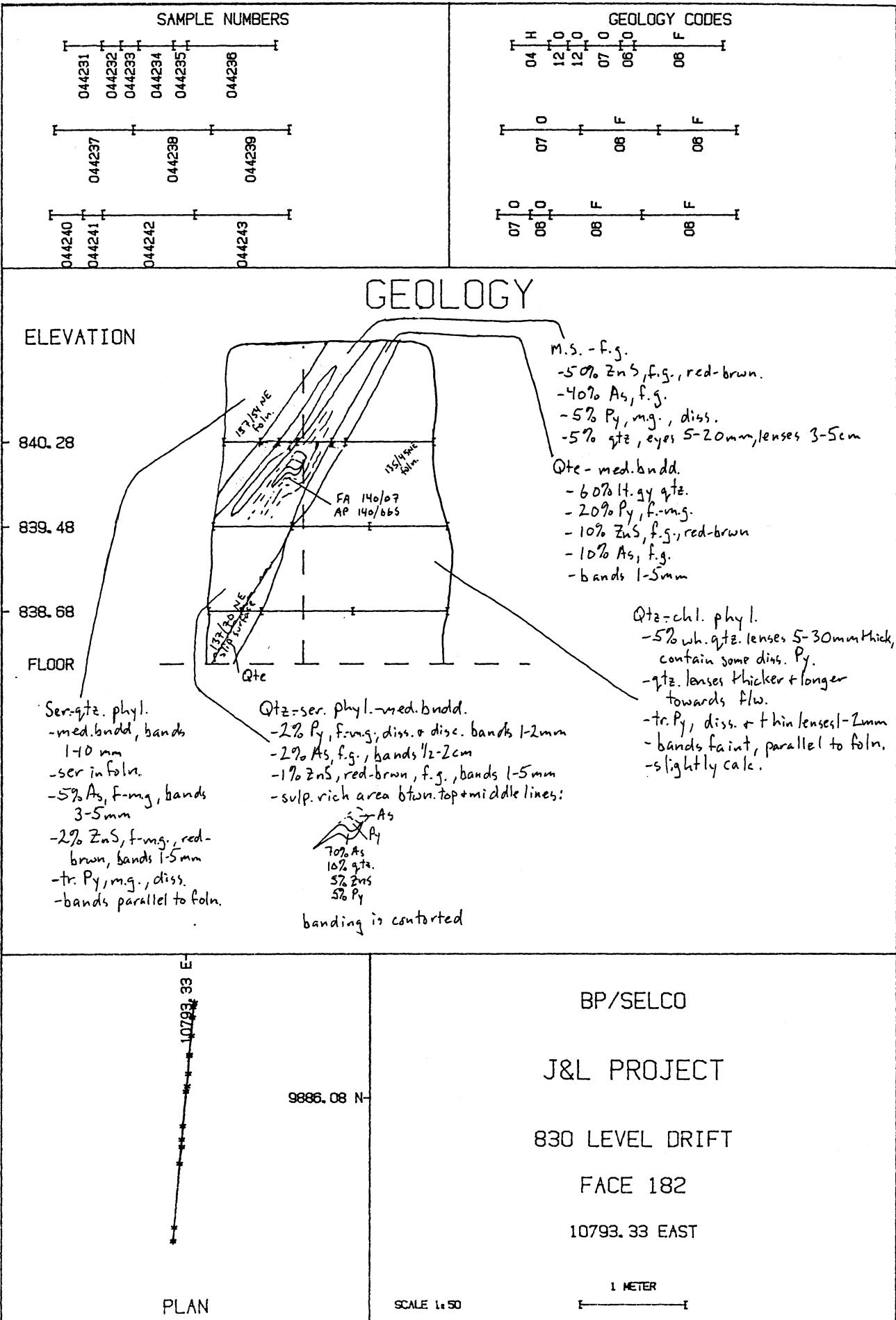
FACE 181

10790.64 EAST

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 182

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44231	.35	.28	4.19	.363	3.1	.3
44232	.18	6.97	17.10	2.040	111.5	6.4
44233	.17	4.34	11.10	3.950	92.0	4.0
44234	.33	.32	.18	7.420	9.0	2.0
44235	.13	1.75	2.32	9.980	47.0	9.9
44236	.84	.03	.04	.130	2.1	.6
44237	.75	.30	.50	.787	6.3	1.2
44238	.74	.01	.02	.078	1.3	1.4
44239	.74	.01	.01	.030	1.7	.3

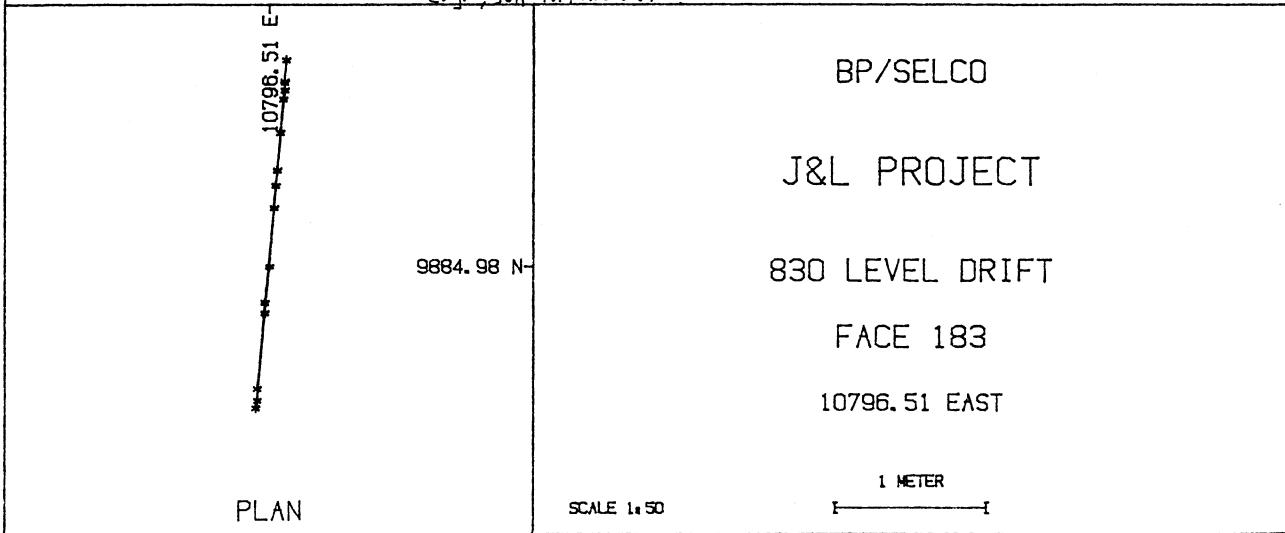
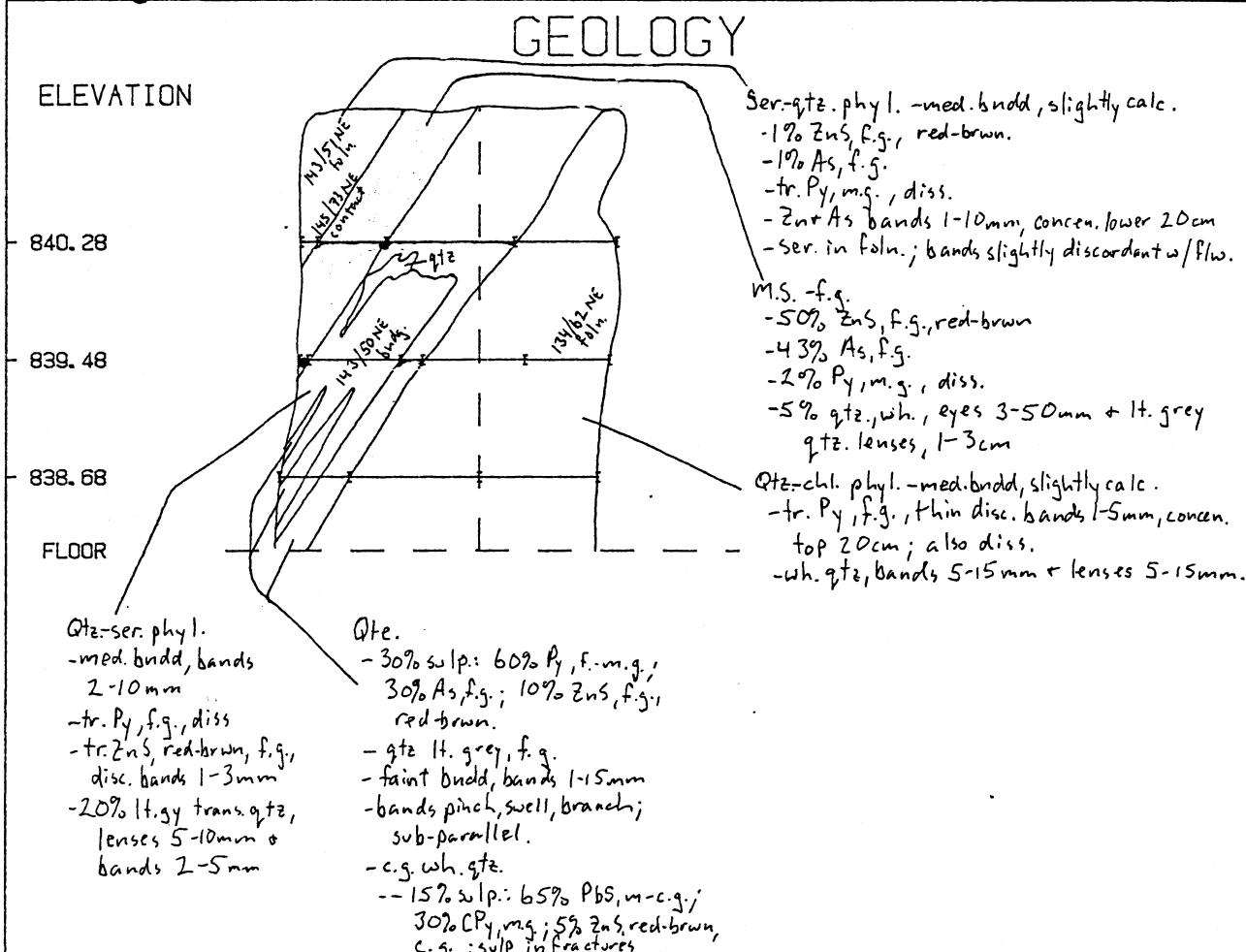
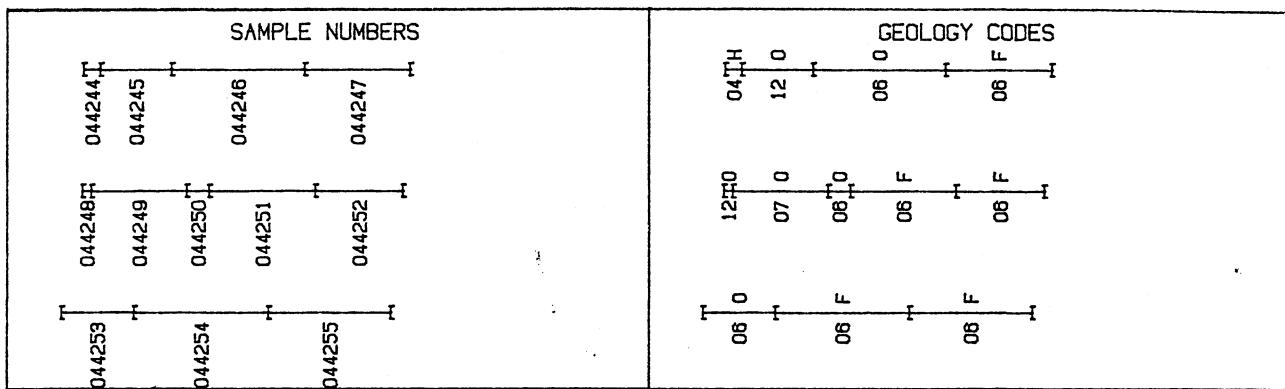


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 183

* * * * *

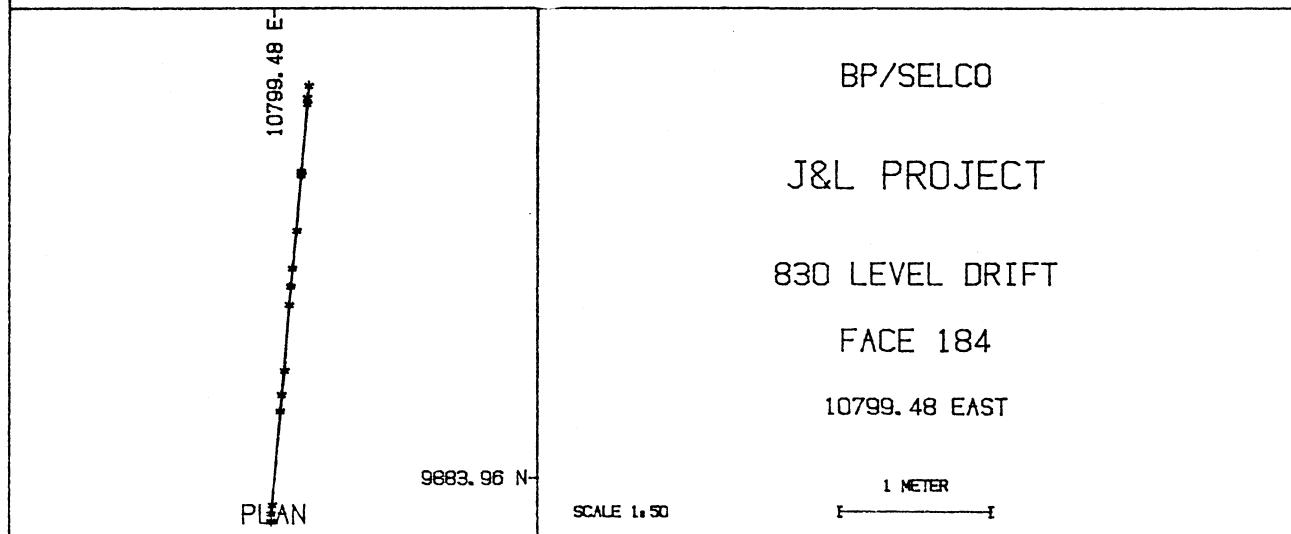
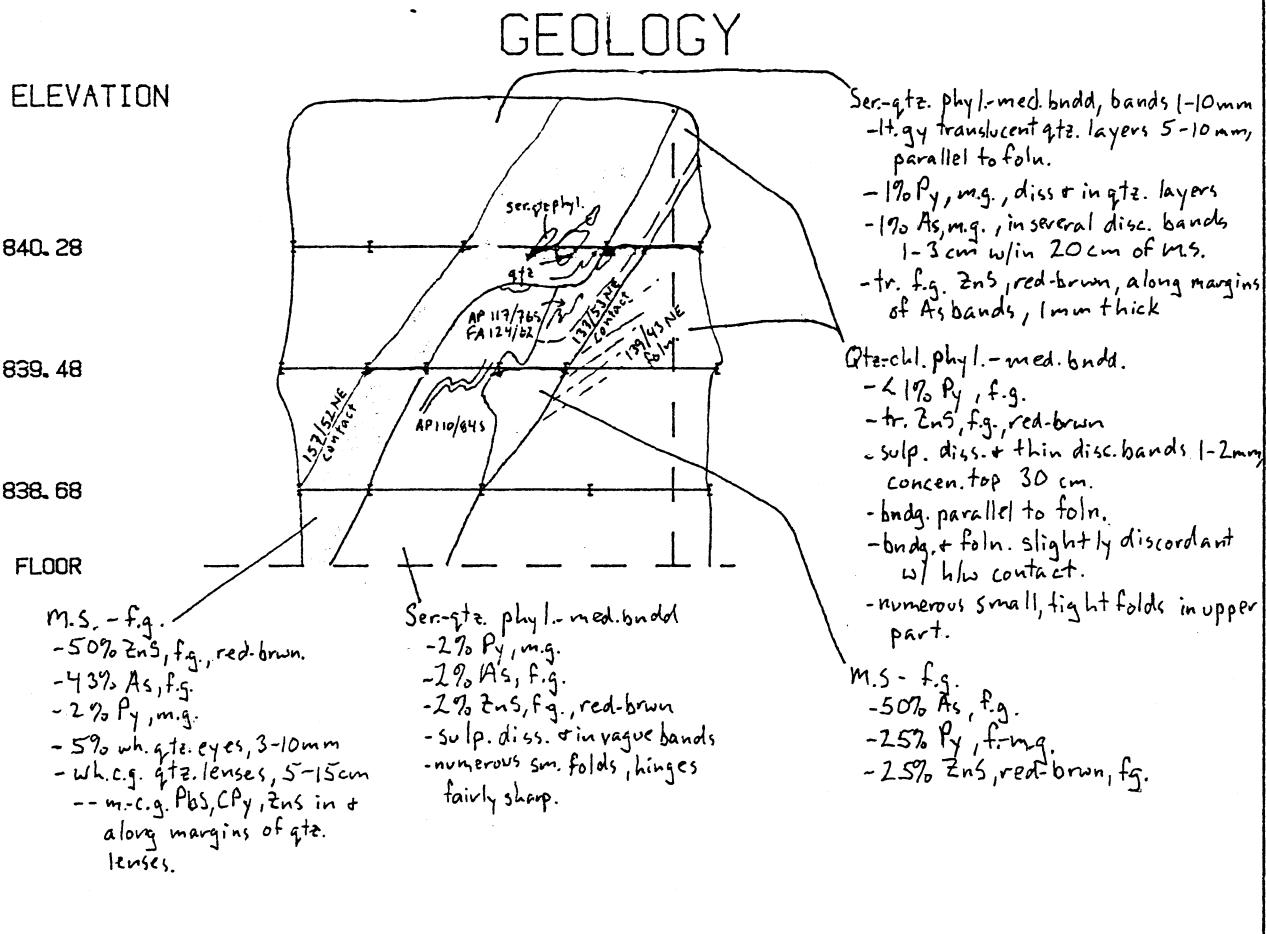
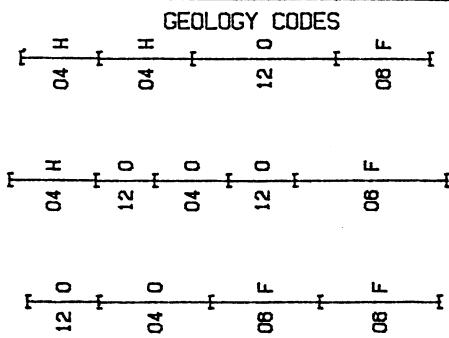
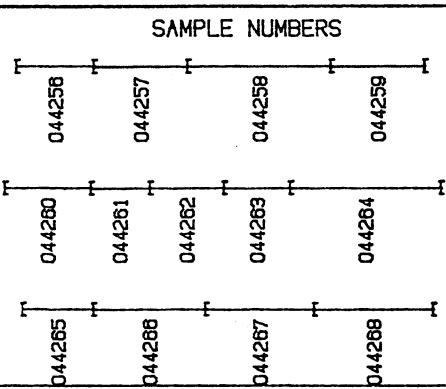
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44244	.11	.81	3.36	.165	18.9	.3
44245	.47	4.84	14.40	1.720	84.4	4.7
44246	.88	.90	.96	6.580	25.3	5.5
44247	.70	.02	.06	.249	1.2	.2
44248	.06	5.92	16.30	2.330	91.3	5.4
44249	.63	1.42	1.03	.616	13.7	.7
44250	.15	.81	1.23	6.380	17.1	6.2
44251	.70	.02	.03	.334	1.9	.8
44252	.58	.01	.02	.132	1.7	.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 184

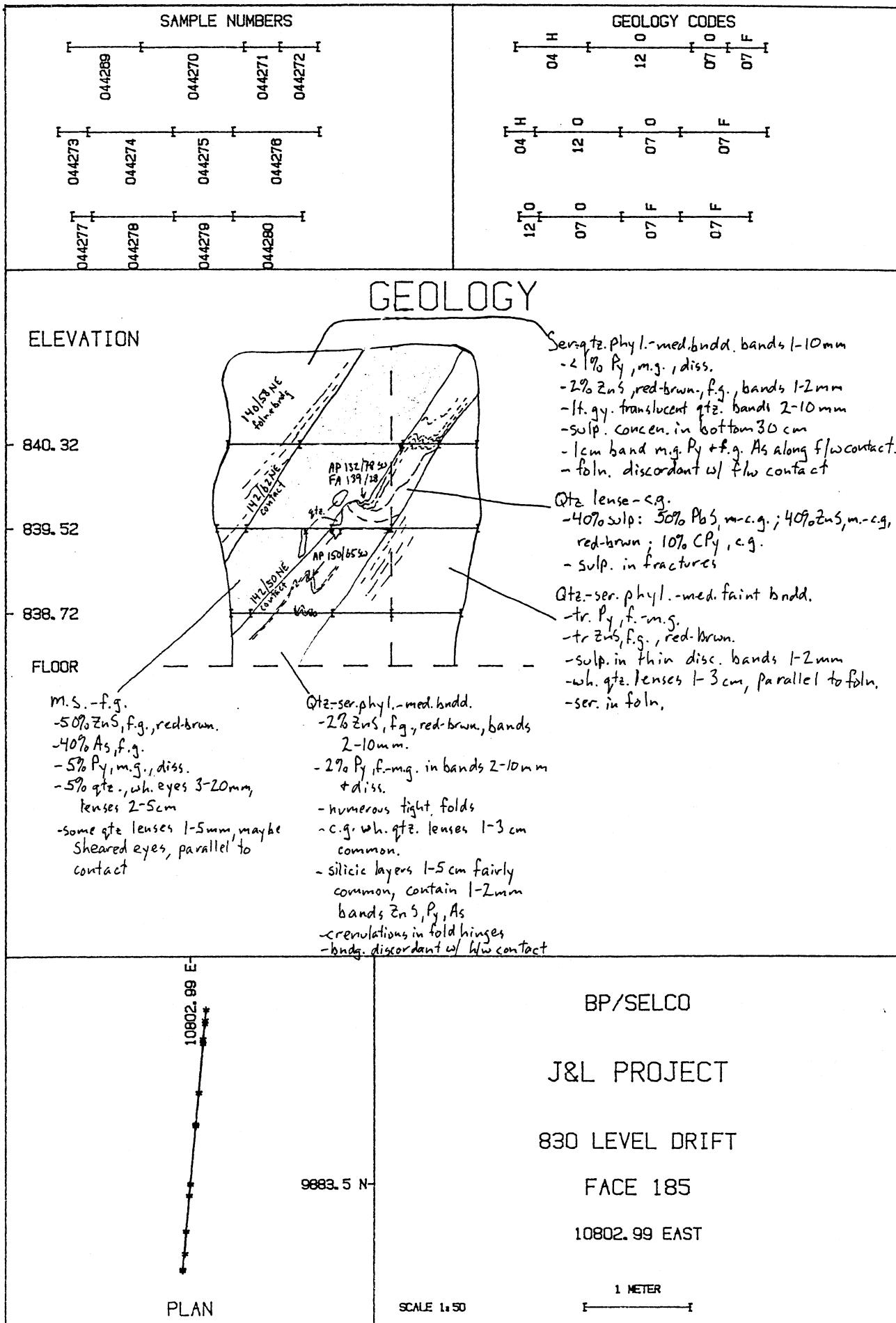
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44256	.51	.01	.02	.071	1.9	.1
44257	.62	.17	.17	2.460	5.9	.3
44258	.95	5.61	13.50	1.520	97.3	4.9
44259	.62	14.90	4.43	.718	333.2	4.8
44260	.57	.23	.16	.775	6.5	.3
44261	.39	5.25	14.60	1.980	74.6	5.6
44262	.49	.11	.17	.278	4.5	.3
44263	.44	1.81	1.54	10.700	45.5	63.5
44264	1.00	.06	.04	.230	2.8	.6



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 185

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44269	.69	.05	.16	.037	2.6	.1
44270	.98	4.12	13.90	1.580	74.0	3.5
44271	.34	4.06	5.22	.667	94.3	1.6
44272	.36	.11	.27	.088	6.0	.2
44273	.28	.10	.72	.138	4.6	.3
44274	.81	3.86	15.80	1.700	74.3	3.8
44275	.57	6.96	5.54	1.790	155.2	6.6
44276	.82	.08	.14	.027	2.5	.2

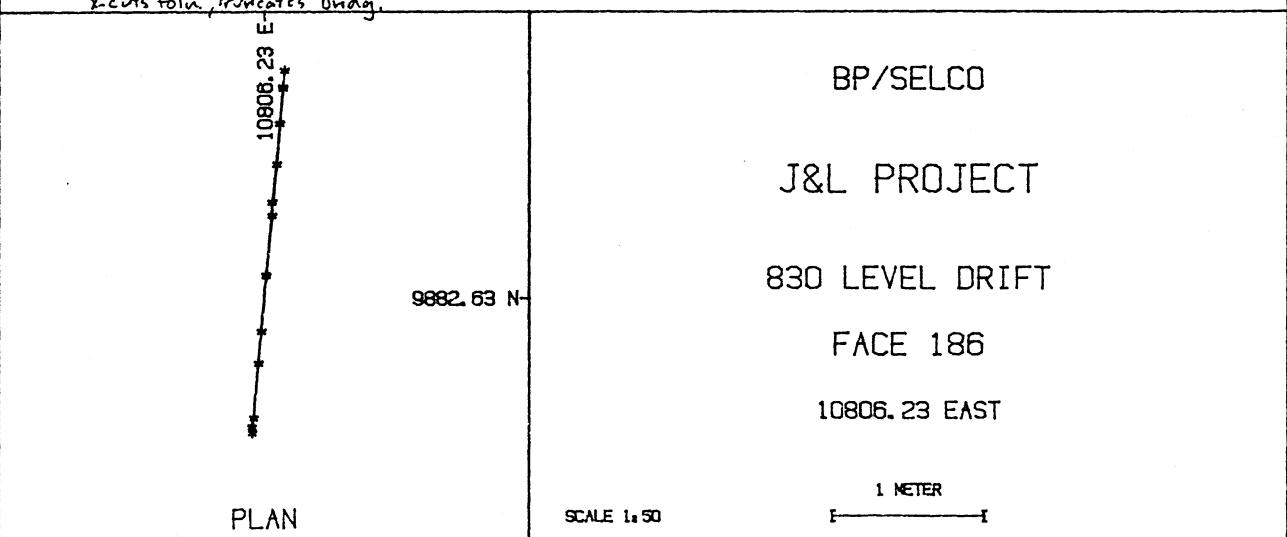
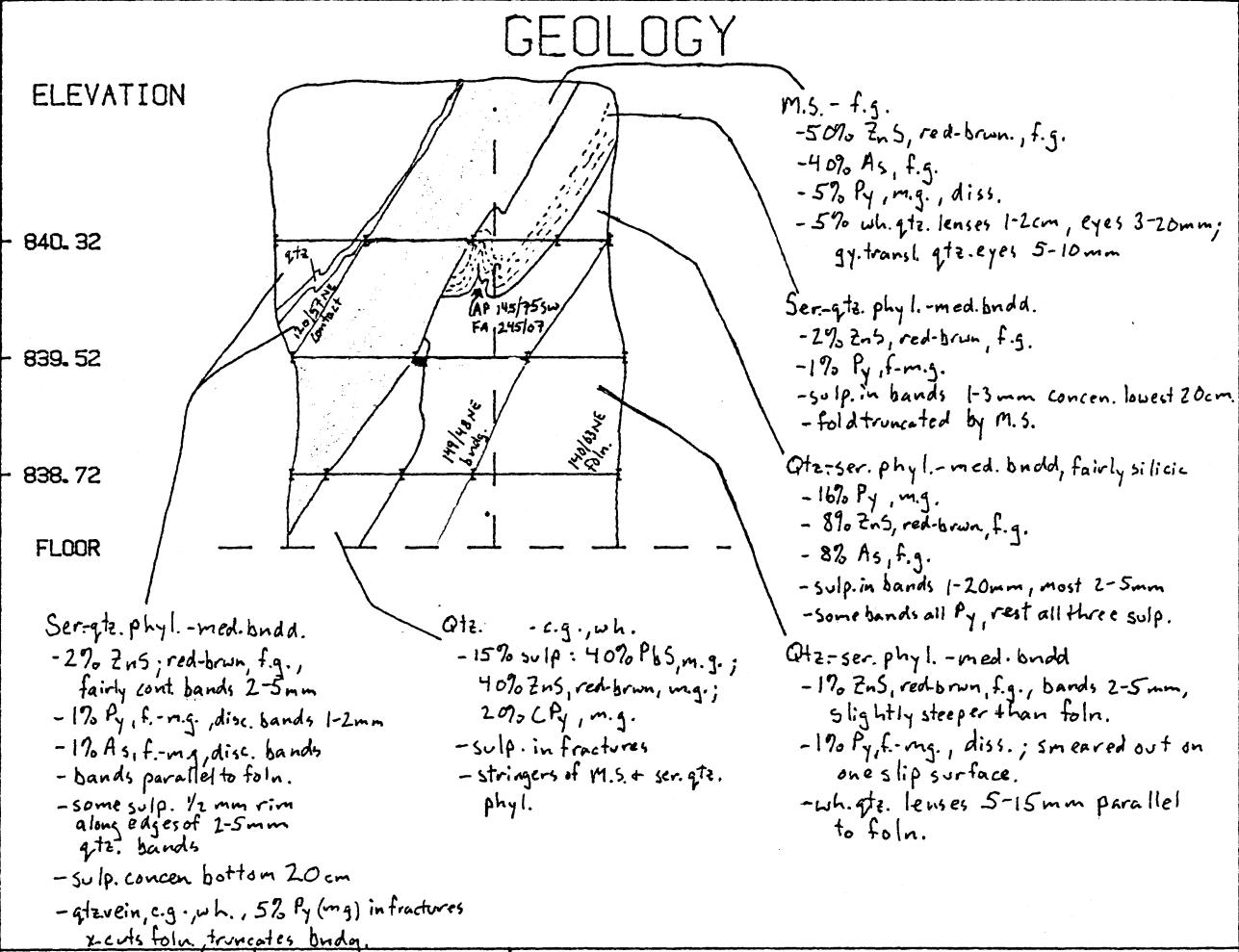
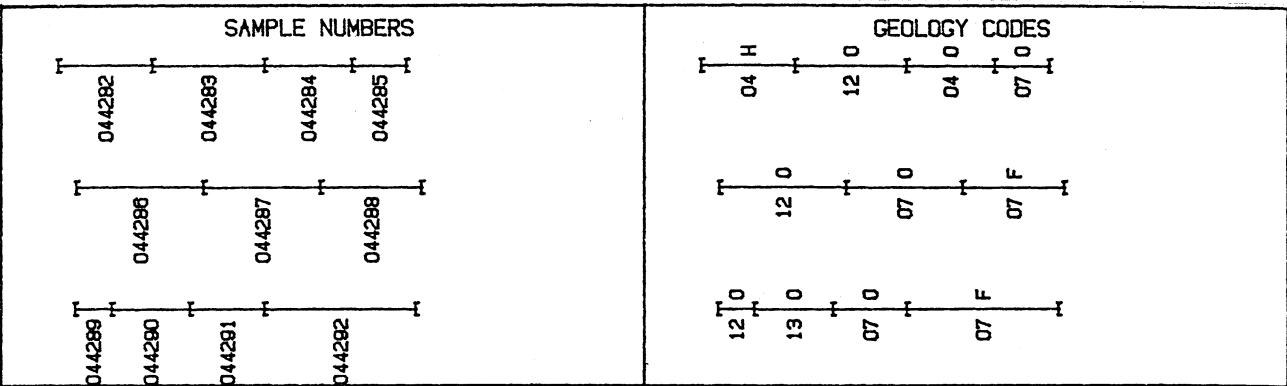


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 186

* * * * *

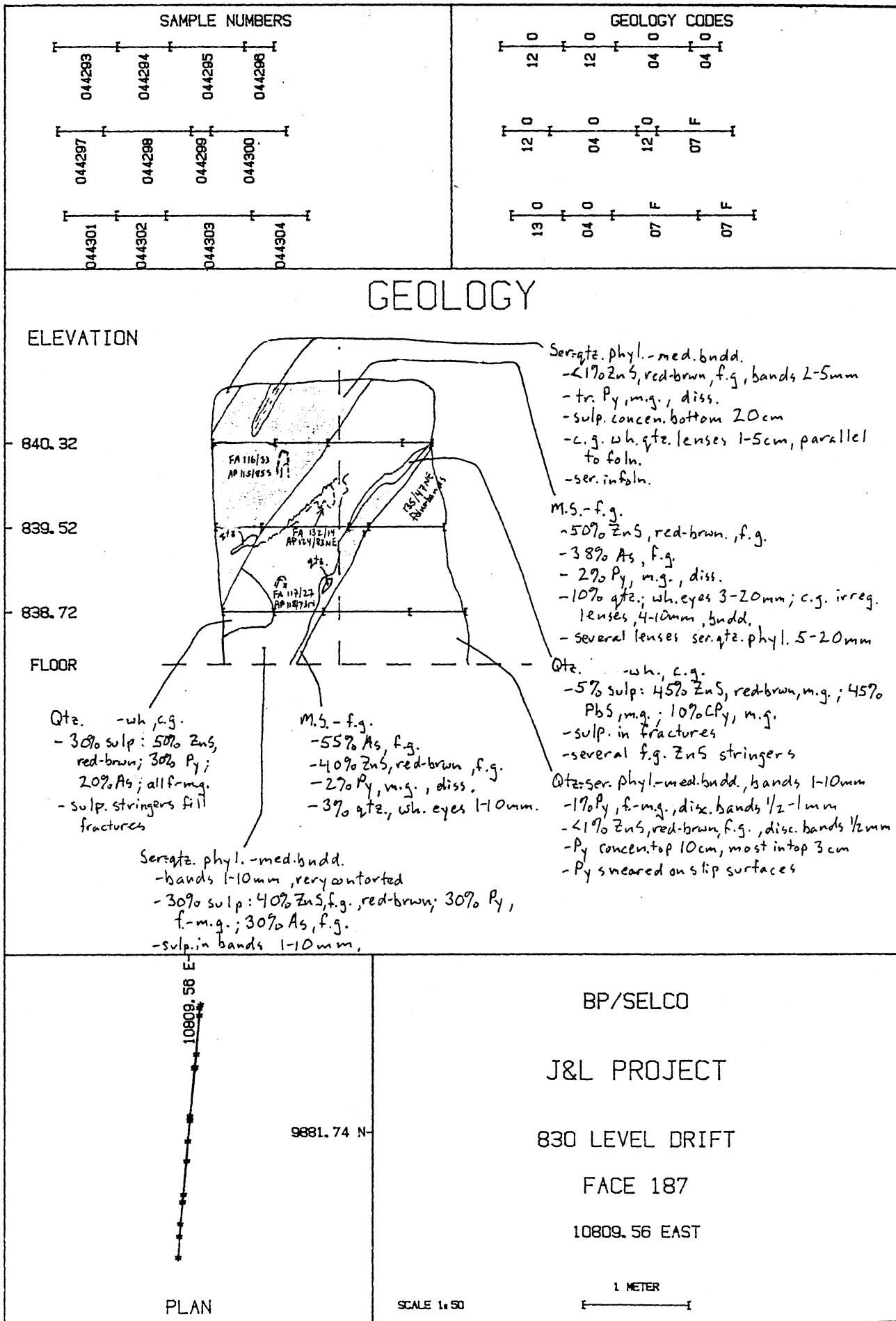
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44282	.62	.04	.13	.018	.6	.1
44283	.74	5.33	16.60	2.080	80.5	5.2
44284	.58	.35	1.51	1.510	7.8	1.1
44285	.36	.98	1.64	4.340	26.7	3.5
44286	.84	4.62	15.30	1.880	90.4	4.3
44287	.77	5.62	5.06	7.080	128.5	8.6
44288	.67	.07	.08	.218	2.4	.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 187

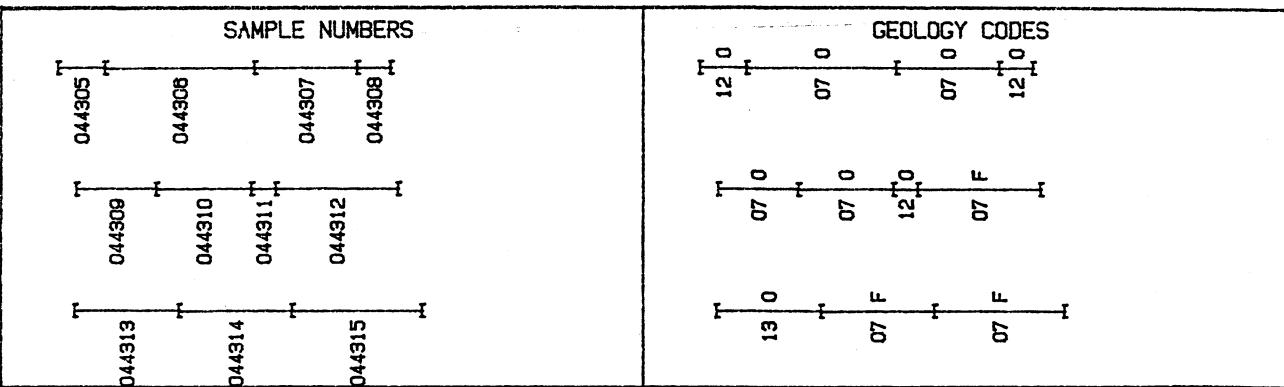
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44293	.60	3.80	12.30	2.660	71.0	5.1
44294	.50	3.62	13.50	1.480	74.4	3.1
44295	.71	.18	1.79	.264	3.9	.2
44296	.28	.18	1.00	1.000	5.5	.7
44297	.44	3.80	12.30	1.540	80.5	3.1
44298	.83	.91	1.94	1.050	17.8	.7
44299	.19	4.20	5.83	4.830	97.7	4.4
44300	.72	.18	.09	2.630	3.2	.9



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

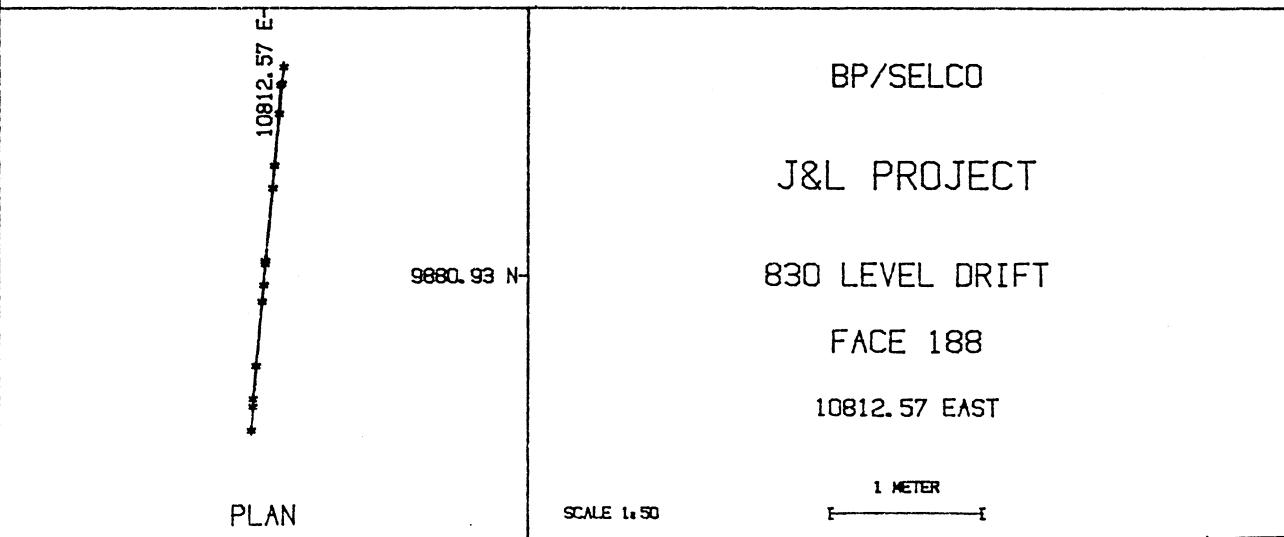
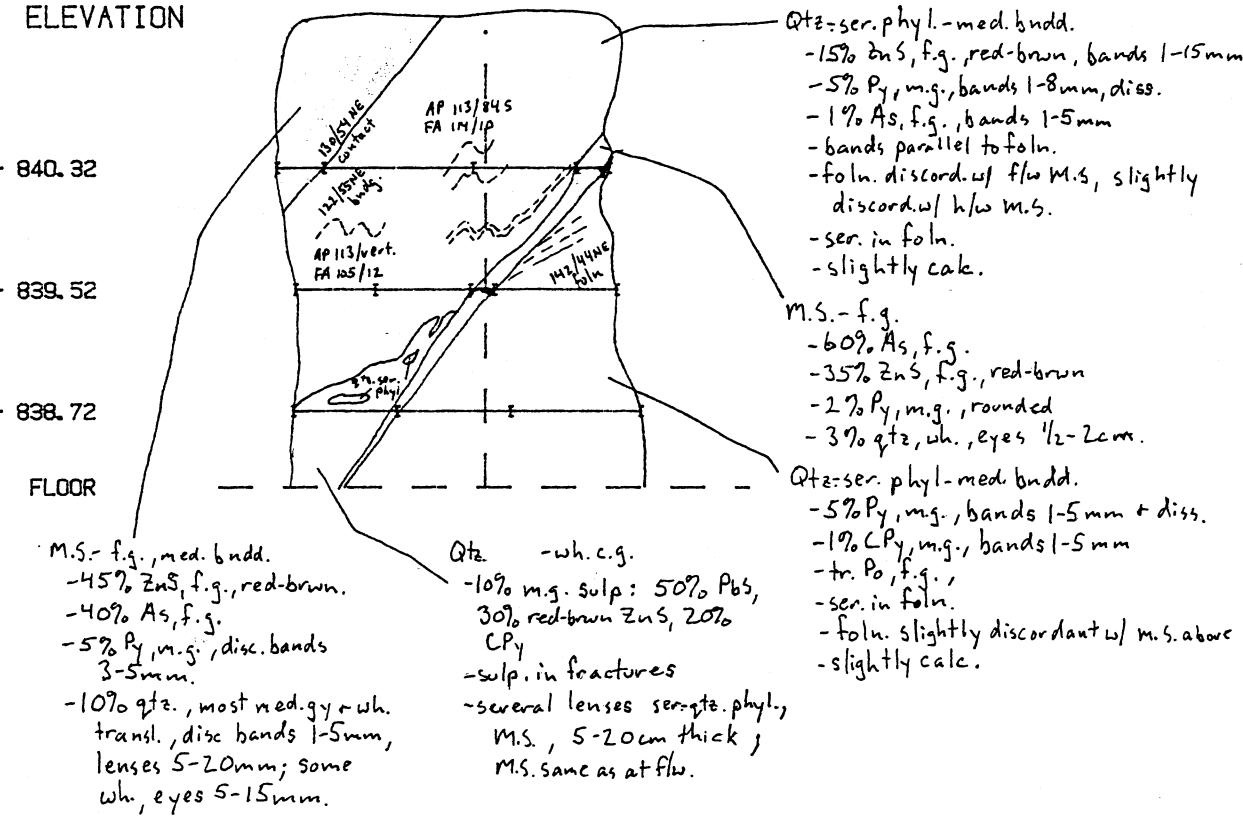
FACE# 188

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44305	.31	4.68	11.70	.961	85.3	3.1
44306	.99	.23	1.58	.649	.8	1.2
44307	.68	.18	.88	.848	.3	.5
44308	.22	5.61	7.30	8.310	117.4	9.4
44309	.53	.31	4.35	.388	6.4	.4
44310	.63	.54	2.46	.595	13.2	.5
44311	.16	4.84	5.06	4.930	111.8	4.7
44312	.81	.10	.10	.518	3.0	.4



GEOLOGY

ELEVATION

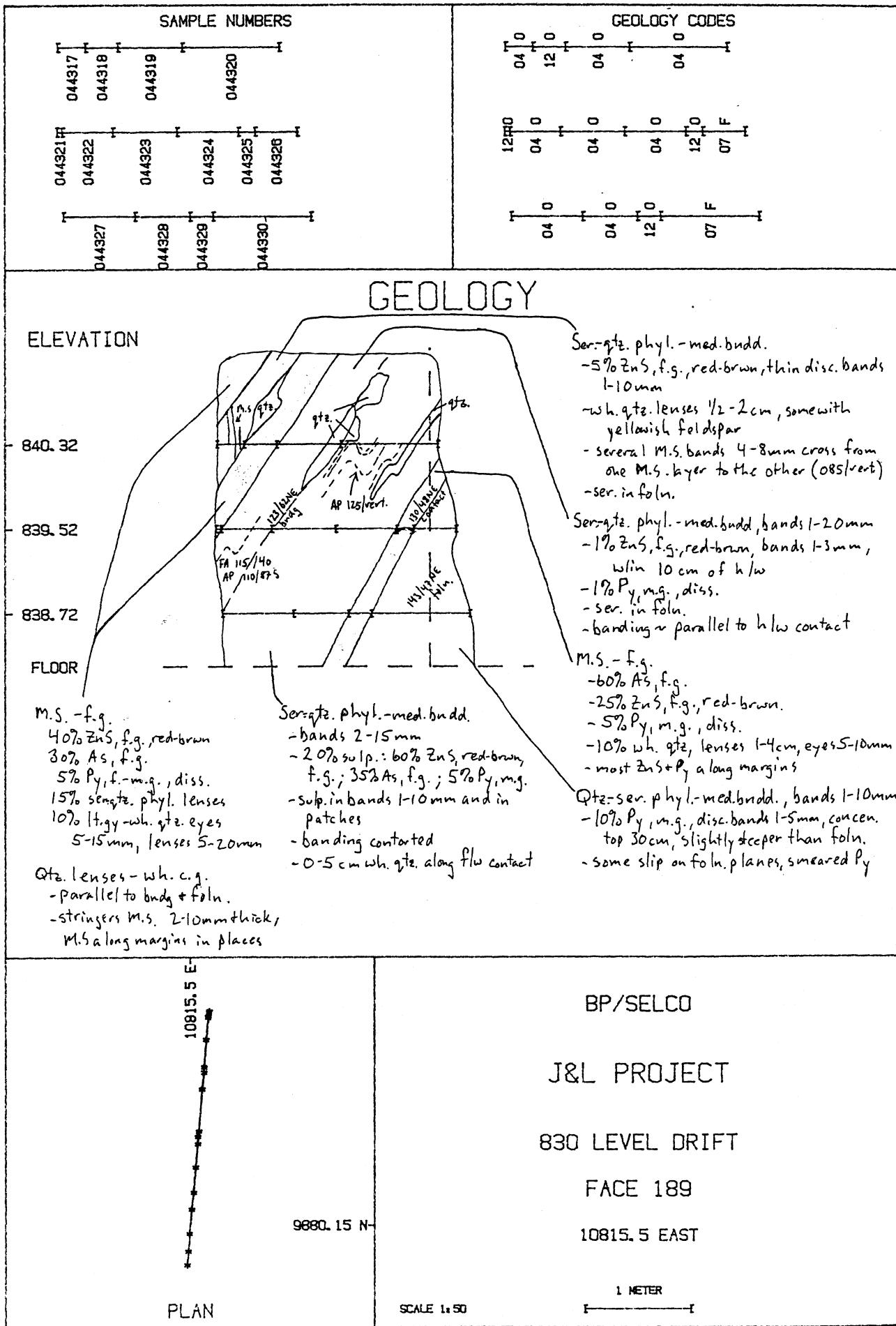


SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 189

* * * * *

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44317	.26	.63	2.13	.262	10.2	.8
44318	.31	2.44	8.52	.478	40.2	1.4
44319	.61	.16	.89	.662	.3	.3
44320	.92	.10	.67	.726	3.0	.4
44321	.05	2.56	8.52	.550	42.5	1.4
44322	.48	.16	.55	.418	.3	.3
44323	.61	.30	1.62	1.230	2.4	1.0
44324	.58	.28	.61	1.500	2.7	.7
44325	.16	4.03	6.72	8.010	76.5	7.8
44326	.40	.06	.08	.128	.3	.3



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

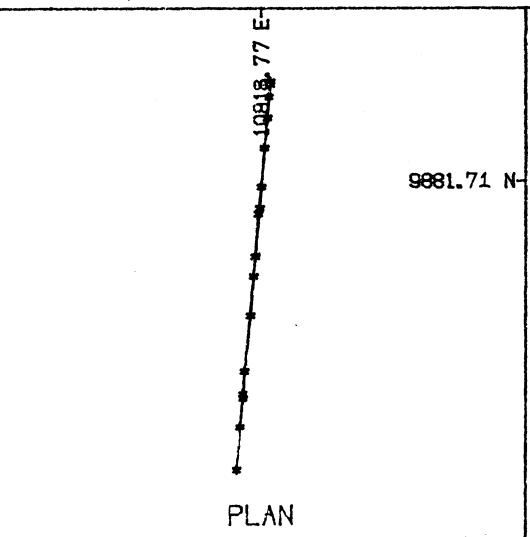
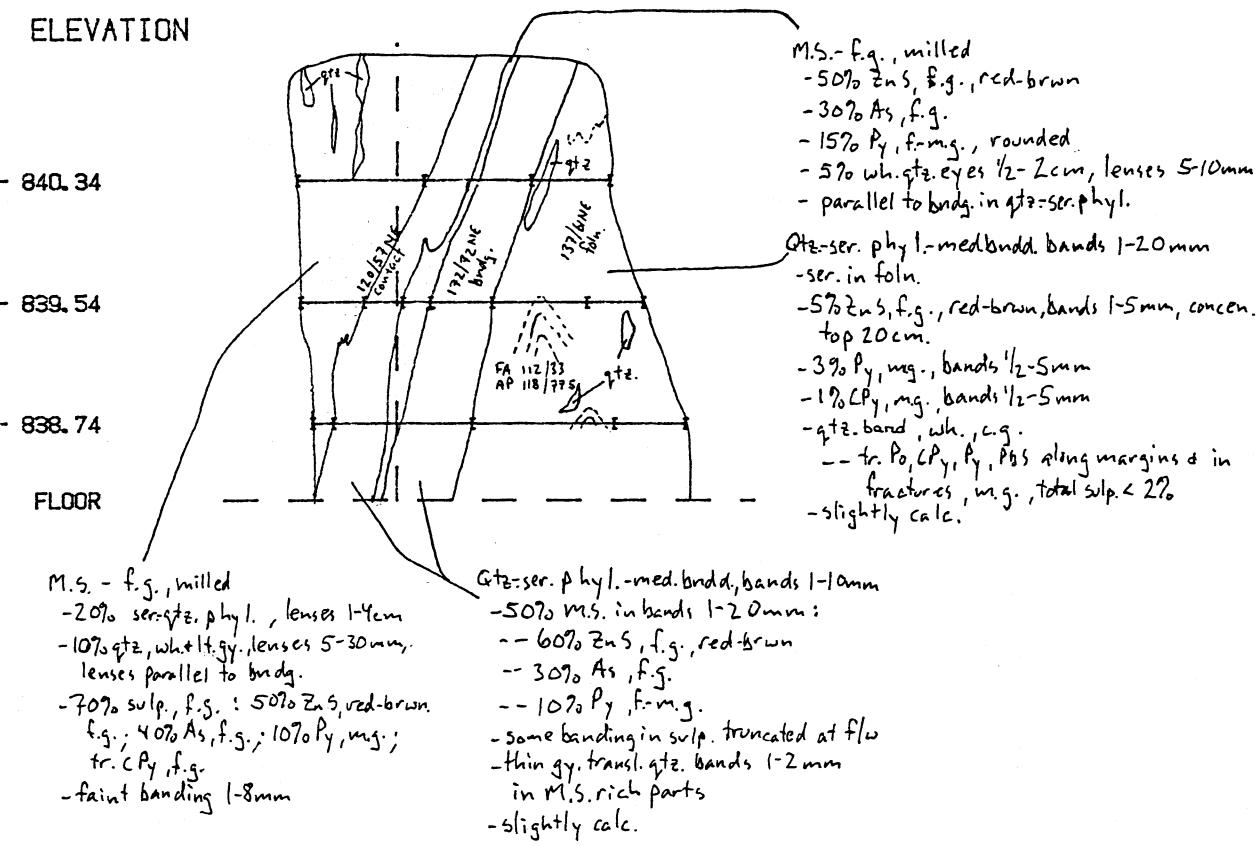
FACE# 190

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44331	.84	2.26	7.44	.738	39.5	1.6
44332	.71	1.84	8.86	3.380	38.5	4.0
44333	.52	.04	.17	.073	13.3	.4
44334	.42	4.24	16.00	2.390	74.0	4.1
44335	.26	.35	.95	1.730	15.7	.7
44336	.18	2.37	10.90	4.580	47.9	4.9
44337	.41	1.08	9.60	.175	22.6	1.4
44338	.63	.06	1.04	.129	1.7	.3
44339	.37	.05	.65	.392	5.9	.3

SAMPLE NUMBERS

GEOLOGY CODES

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 190

10818.7 EAST

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

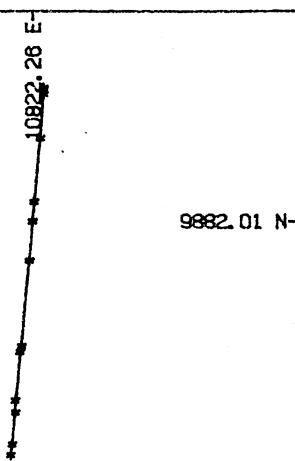
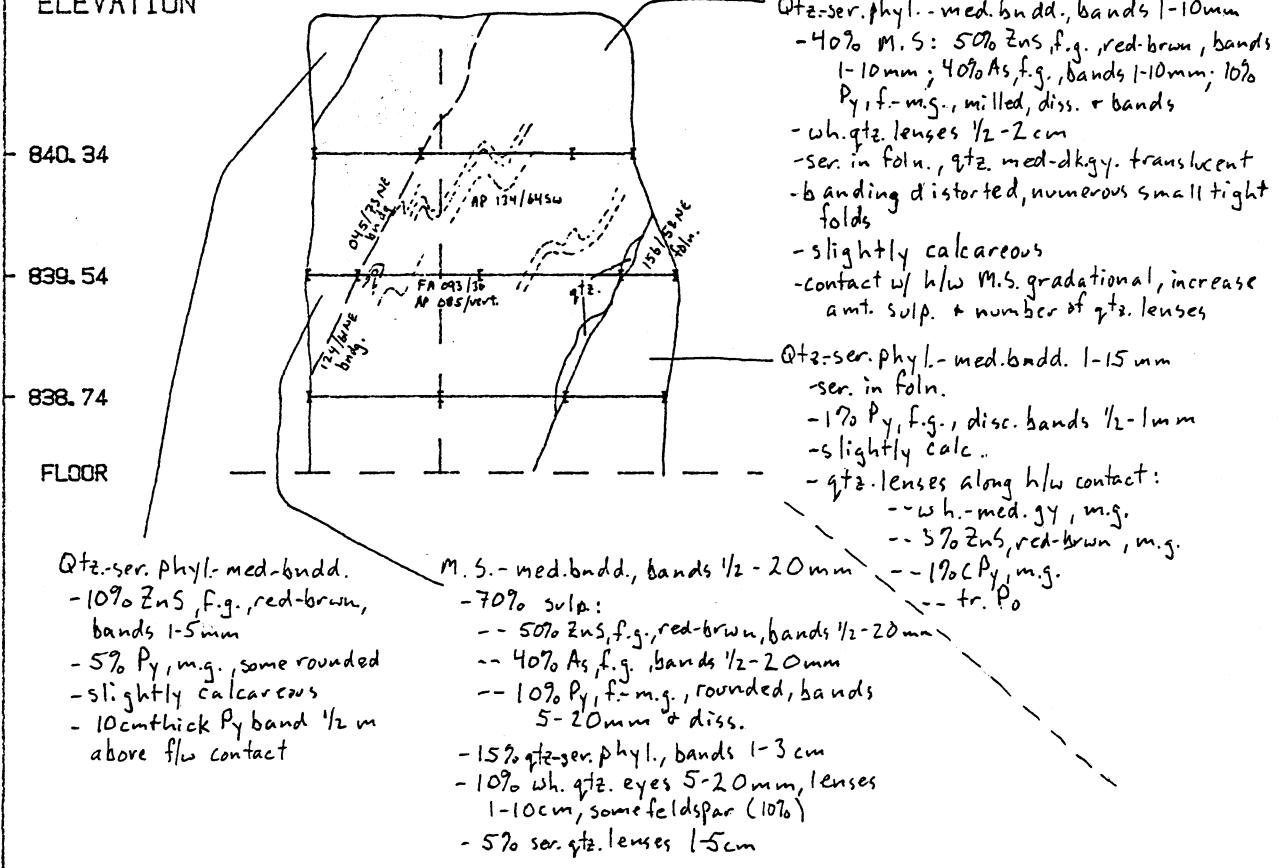
FACE# 191

SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44344	.71	4.16	16.60	.647	68.5	1.4
44345	1.00	.37	3.58	.432	5.9	.3
44346	.40	5.60	15.80	.030	90.6	.6
44347	.33	4.09	15.30	.757	73.1	1.6
44348	.81	2.31	7.81	.907	36.9	.8
44349	.93	1.44	5.00	.742	21.9	.7
44350	.36	.06	.25	.011	.7	.1

SAMPLE NUMBERS			GEOLOGY CODES		
044344	044345	044346	12 0	07 0	07 0
044347	044348	044349	12 0	07 0	07 F
044351	044352	044353	07 0	07 0	07 F

GEOLOGY

ELEVATION



BP/SELCO

J&L PROJECT

830 LEVEL DRIFT

FACE 191

10822.26 EAST

1 METER

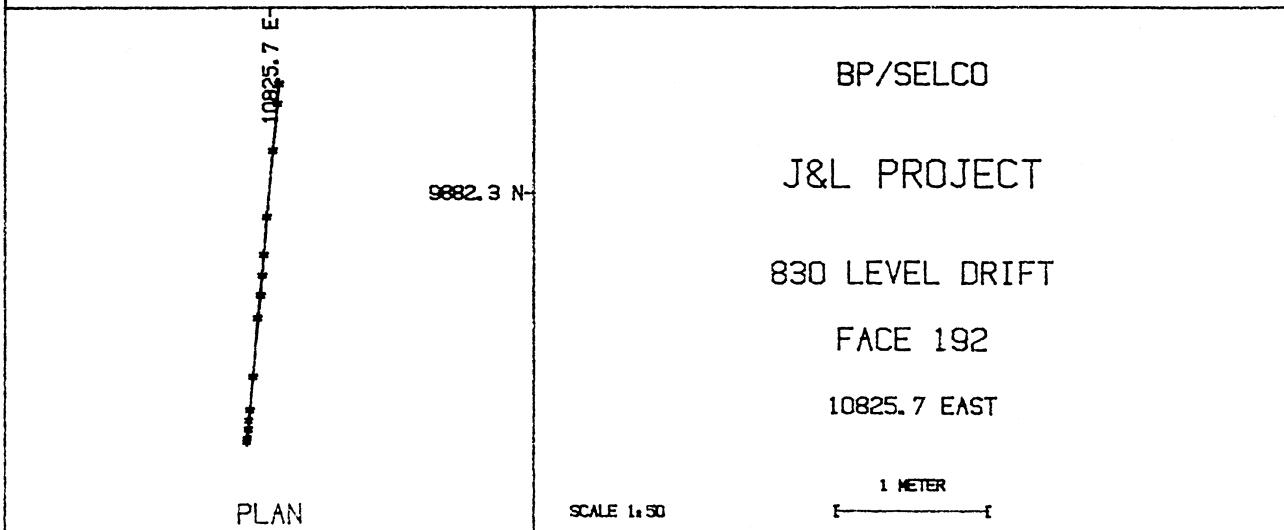
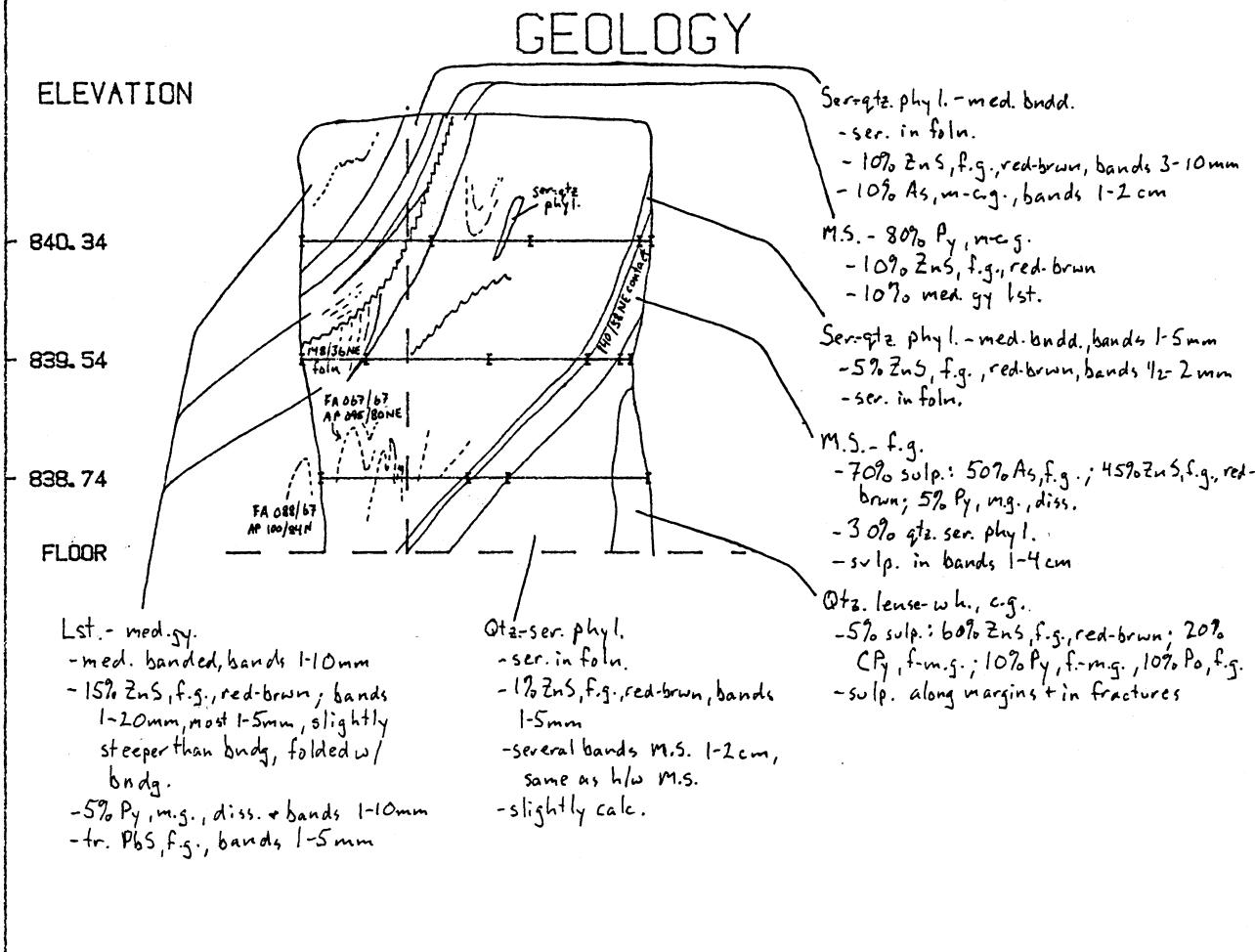
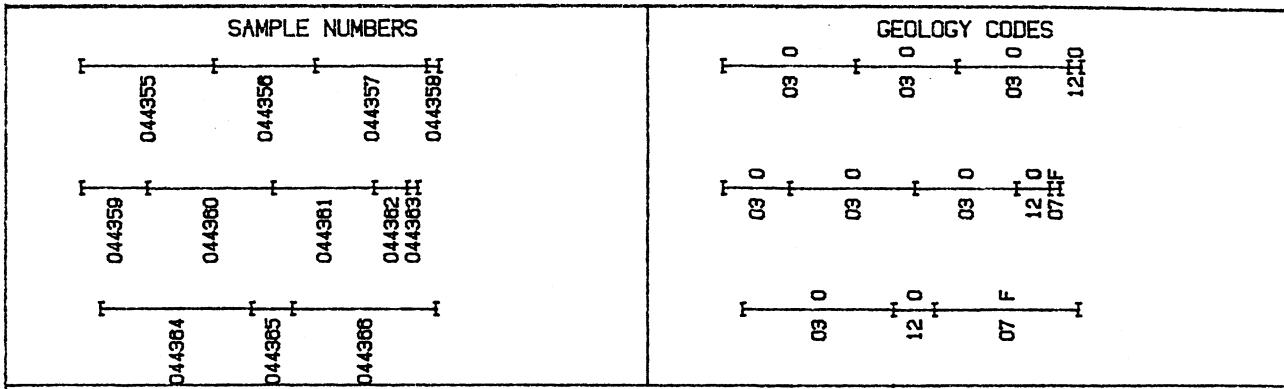
PLAN

SCALE 1:50

SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 192

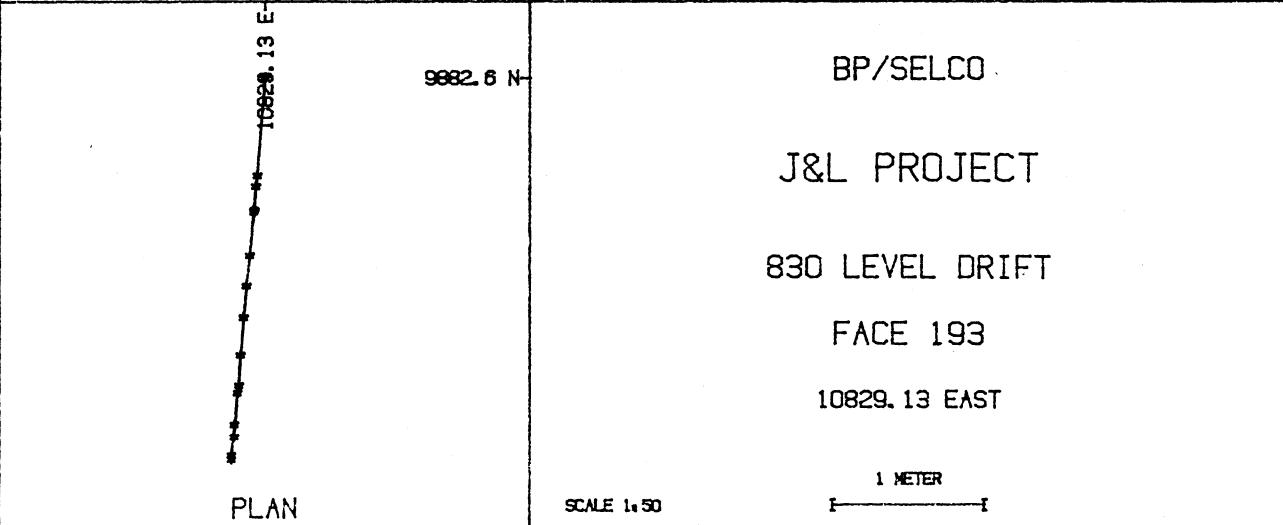
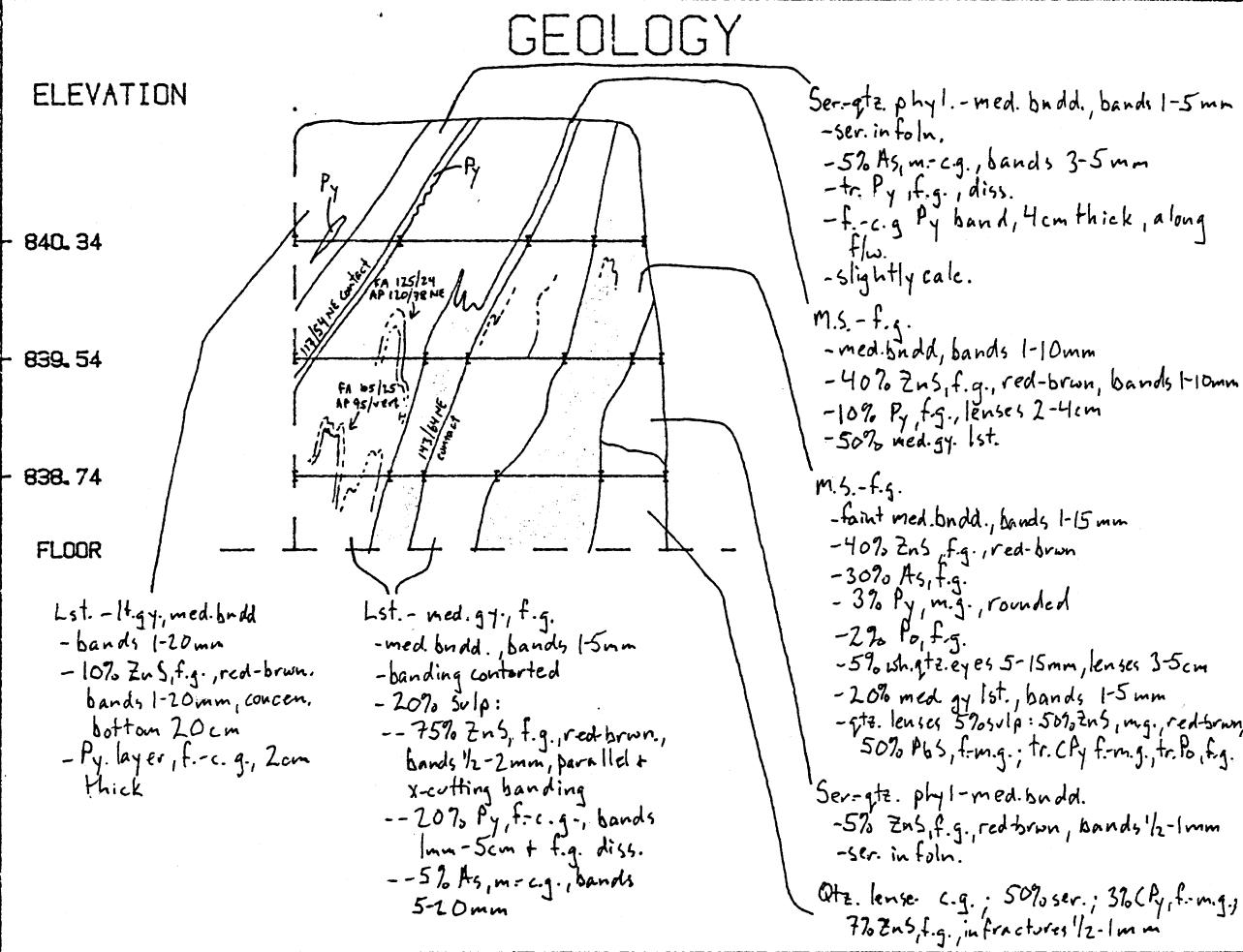
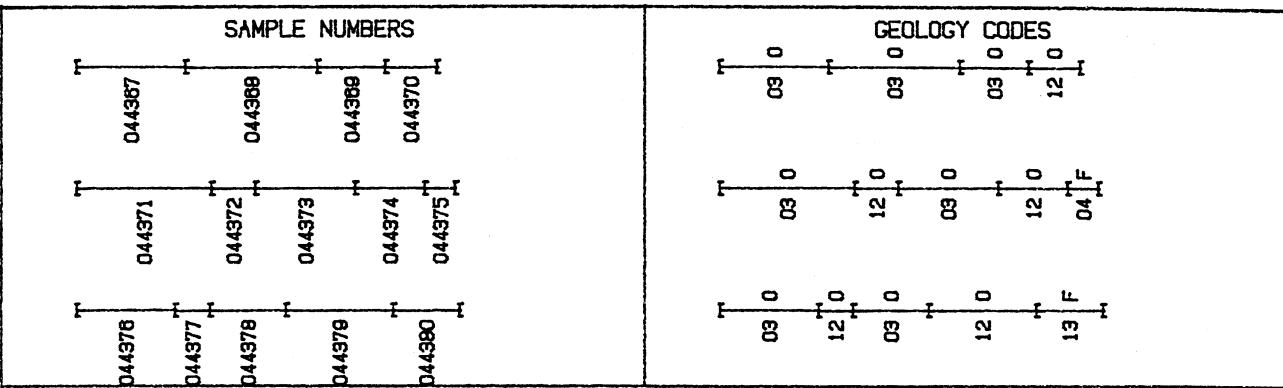
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44355	.88	2.16	4.52	2.920	35.2	1.1
44356	.67	.70	1.88	.124	11.4	.2
44357	.74	.70	1.59	.862	13.7	.1
44358	.08	2.62	12.50	2.970	55.1	2.5
44359	.44	.91	5.40	.129	14.4	.1
44360	.83	1.39	3.05	.039	19.7	.2
44361	.67	.78	3.03	.834	11.1	.5
44362	.22	.76	7.43	.708	20.5	.7
44363	.07	.12	.34	.208	6.6	.2



SELCO DIV. - BP RESOURCES J & L PROJECT, B.C.

FACE# 193

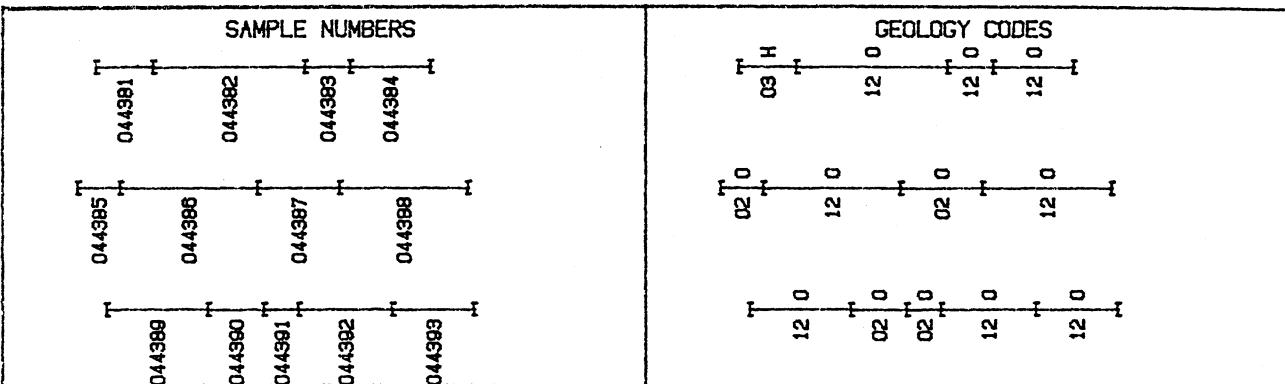
SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44367	.72	.68	2.40	.670	10.5	.5
44368	.87	3.19	5.92	.029	16.9	.2
44369	.45	.53	5.54	.417	8.7	.2
44370	.34	3.29	11.80	2.880	54.5	2.4
44371	.89	.99	3.53	.644	14.8	.3
44372	.29	1.06	13.50	.088	15.1	.7
44373	.66	1.03	5.22	.039	18.2	.3
44374	.46	1.92	8.22	.449	38.7	.4
44375	.20	.10	.56	.059	3.4	.1



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SAMPLE NUMBER	WIDTH (m)	LEAD (%)	ZINC (%)	ARSENIC (%)	SILVER (g/mT)	GOLD (g/mT)
44381	.38	2.08	11.60	.358	33.9	1.1
44382	1.00	1.07	29.60	.210	20.3	.3
44383	.30	.12	30.50	.026	12.3	.7
44384	.53	2.14	12.70	.559	36.2	1.5
44385	.28	.62	3.01	.204	9.5	.8
44386	.91	.46	30.10	.010	9.3	.3
44387	.54	.29	19.20	.009	5.9	.3
44388	.85	3.62	11.50	.703	61.5	1.6



GEOLOGY

ELEVATION

