

RECEIVED GOVERNMENT AGENT SALMON ARM
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ASSESSMENT REPORT

GEOCHEMICAL SOIL SURVEY

LEM 2 MINERAL CLAIM

TAWEEL LAKE AREA Kamloops Mining Division NTS: 92P/9w

Lat: 51 deg. 37'N - Long: 120 deg. 15'W

For:

Zeo-Tech Enviro Corp. Ste. 2300 – 1066 West Hastings St. Vancouver, BC V6E – 3X2

By:

John Jenks - P.Geo. (BC) Salmon Arm, BC V1E - 3C7

June 23, 2003

GEOLOGICAL SURVEY BRANCH ASSECTMENT PEPORT

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SUMMARY:

The author was retained by Mr. Ray Paquette, president of Zeo-Tech Enviro Corp to initiate soil geochemical coverage of the Lem 6 claim in an ongoing effort to explore and eventually cover the entire claim area.

The Lem 2 is a four-post, six unit mineral claim totaling 150 hectares and is wholly-owned by Zeo-Tech.

Situated 24 kilometres NNW of Little Fort, BC the claims may be accessed by 30 km of bad road to Taweel Lake and the southern claim boundary.

The property has undergone poorly-documented exploration in the past including trenching and diamonddrilling.

Located on the eastern margin of the Quesnel Trough the claims are primarily underlain by Triassic-aged flysch sediments, late Jurassic volcanics with at least one monzonite dyke exposed of apparent Cretaceous age. Previous trenching has revealed gold-rich massive sulphides (pyrite, pyrrotite, chalcopyrite, galena and sphalerite) in two separate locations.

During the September 2002 programme the author collected 50 soil samples at 25 metre intervals along lines spaced 200 metres apart. The samples were submitted to Acme Analytical Laboratories in Vancouver where they were subjected to 35 element ICP and MS analysis.

Results indicate the presence of three anomalous areas warranting subsequent follow-up. Recommendations are made for extending coverage, prospecting and a trial EM-16 survey at an estimated cost of \$10,000 to \$15,000.

Expenditures on the property over the course of the survey totaled \$2,949.34.

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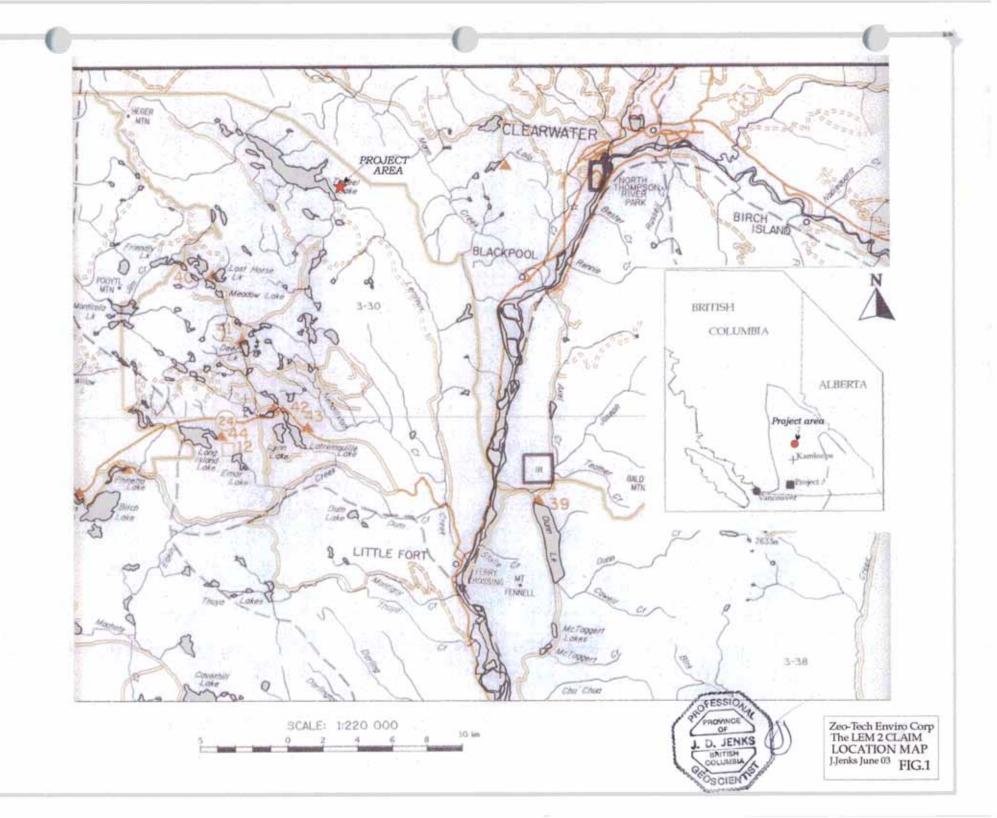
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1. INTRODUCTION:

a) TERMS OF REFERENCE:

Jenks was retained by Mr. Ray Paquette, President of Zeo-Tech Enviro Corp., to initiate soil geochemical coverage of the claim area in order to satisfy assessment requirements and outline areas for possible followup. It is anticipated that coverage would be ongoing in subsequent years and progressively extended to eventually cover the entire claim area.

b) LOCATION AND ACCESS:

The Lem claims are situated 24 air kilometres NNW of Little Fort, B.C. and 18 kilometres west of Clearwater, B.C. During the project they were accessed by 4-W drive road from Little Fort as follows:

Odometer (Kms)	Description
0	Junction of highways #5 and #24 at Little Fort. Take highway #24 west from Little Fort.
3.7	Turn left onto Lemieux Creek logging road.
17.4	Take left fork.
19.6	Take left fork.
20.8	Left turn onto private property. Stay right.
24.3	Cross over Fourteen Mile Creek.
24.6	Left turn to Tinthohtan Lake. Stay right.
29.0	Arrival at Taweel Lake/Nehalliston Fishing Camp situated within claim area.

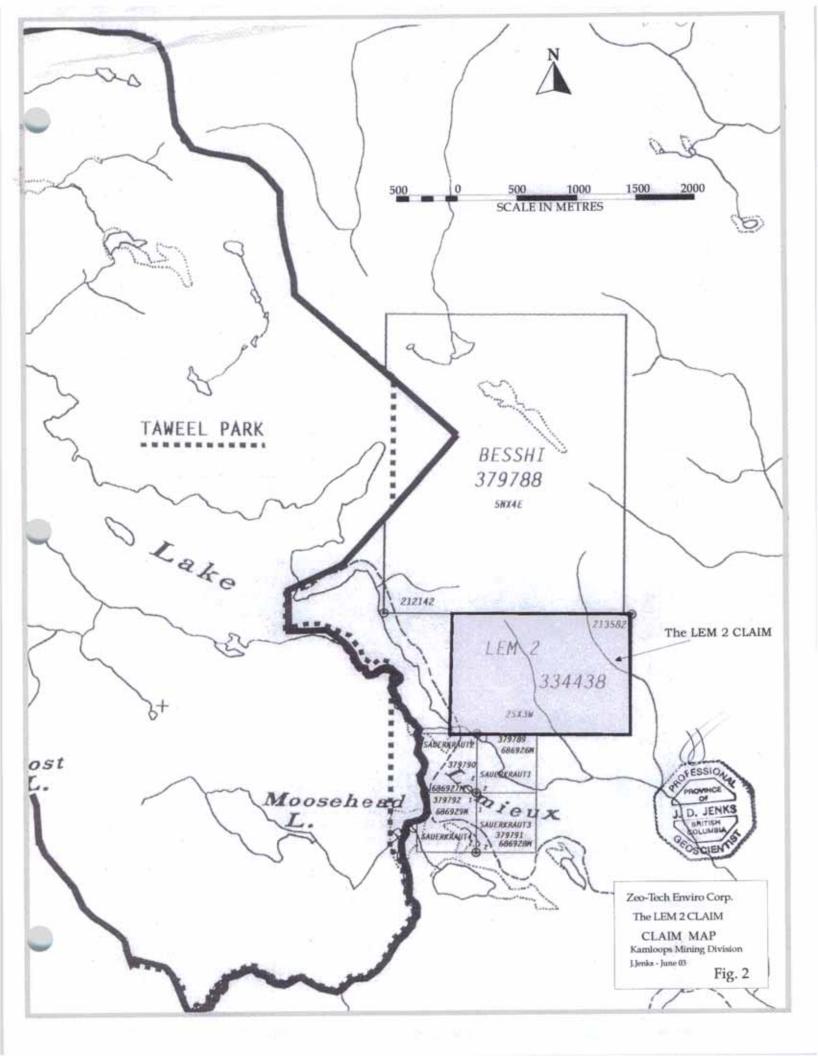
The final eight kilometres of road are extremely rough requiring a four-wheel drive vehicle and a full hour of driving time. The trip from Little Fort to Taweel Lake requires 1 ½ hours to complete. Aside from a rough road used by cottagers skirting the southeast side of the lake access within the claim area is primarily by foot.

c) TOPOGRAPHY, VEGETATION AND CLIMATE:

The claim area consists of rolling, hilly upland terrain with elevations ranging from 1,000 to 1,250 metres occasionally dissected by steep northwesterly-rending fault draws. Well-timbered, the forest cover is thick, sub-mature with a mixture of conifers (spruce, hemlock, balsalm fir, jackpine) up to ¼ metres in diameter. Logging activities progressively encroach upon the area from the northeast. Two permanent streams traverse the claim area – Lemieux Creek draining Taweel Lake at the south end and a small un-named creek emptying into the lake at the same south end. Snow may arrive anytime after mid-October and remain until the end of April to mid-May.

2. - LAND TENURE:

The Lem 2 is a four-post mineral claim comprised of six contiguous units situated adjacent to the southeastern portion of Taweel Lake. Reduced in 1999 from its original sixteen unit size the bloc now totals 150 hectares. Details are as follows:



<u>Claim Name</u>	No, of Units	Record Number	Tag #	Old Anniversary Date	New Anniversary Date
Lem # 2	6	334438	213582	March 25,2003	March 25,2005

The claims are held by Zeo-Tech Enviro Corp., the owner of record.

With the exception of the southeastern shoreline most of the land surrounding Taweel Lake has been withdrawn from staking and made into a provincial park. The dozen or so summer cabins and two fishing lodges on this southeastern corner place the area into an environmentally sensitive category.

3 - HISTORY OF EXPLORATION:

Poorly documented exploration activity within the the claim area extends back to at least 1925. Collapsed remnants of a short exploration drift driven into a massive sulphide zone in the Lemieux Creek headwaters area appears to date to this period. A series of trenches and test pits located one kilometre to the northeast exposing massive sulphides and secondary oxidation minerals were likely dug at this time or slightly later.

A previous claim owner during the 1980's, Sim Jutras, had a portion of the claim area gridded, soil sampled and surveyed by ground magnetics. Several promising anomalies were never followed up.

Peppa Resources/P.Lieberman drilled three short diamond drillholes in the Lemieux Creek area during 1988, with encouraging results. At least two other holes drilled into the same area prior to Peppa's programme are undocumented, their date of completion uncertain.

Previous sampling in the Lemieux Creek showing and from the trench areas to the northeast has reported values to 1.237 opt gold, 13.7% zinc and 12,225 opt silver.

Assessment work carried out by the author for the previous owner, Forefront Ventures during 1994, included rock sampling and reconnaissance soil sampling. High gold and base metal values indicated by previous workers were substantiated. A partial grid was cut and magnetometer surveyed during February 1996. During 1997 both grid and the ground magnetic coverage were extended. The property was partially mapped in 1999 for the subsequent owner, Canadian Mining Co., the forerunner of Zeo-Tech Enviro Corp.

On a more regional basis a number of programmes were carried out between 1960 and 1990. Anaconda conducted a geochemical survey during the 1960's on a massive sulphide zone near Friendly Lake thirteen kilometres to the west while Falconbridge Nickel explored for molybdenum in pegmatites and quartz veins slightly east of the Lem claims. Immediately south of the Lem 2 claim Amax conducted an extensive programme directed towards molybdenum in and around a small granitic stock which culminated in diamond drilling during 1980. During the past five years trenching and drilling activities have been undertaken by Christopher James Gold Corp. and Cassidy Gold a few kilometres to the west.

Within the general North Thompson area a number of gold sulphide properties and showings are present. These include the Little Fort, Nehalliston Creek, Lakeview, Diamac, Eakin Creek placer, Silver Lake, Chu Chua, Queen Bess, Windpass and Samatosum properties; the latter three have undergone previous production.

4. GEOLOGY:

a) REGIONAL:

The claim area is situated at the eastern margin of the Quesnel Trough near the the Shuswap Metamorphic Complex. Within a swarm of northwesterly and north-northwesterly trending block fault splays the area is underlain by folded and block-faulted late Paleozoic, Mesozoic and early Tertiary volcanic, sedimentary and granitic rocks. Much of the claims are covered by glacial drift and loess, especially around the perimeter of Taweel Lake, and rock exposure is limited to Lemieux Creek, the banks of ravines and gullys and the tops of ridges and hills.

The predominant rock type is a Triassic-aged argillite commonly with more siliceous horizons, layers and lenses. Northwesterly and north-northwesterly trending fault linears are readily discernable on air photographs and manifest themselves on the ground as narrow ravines and gullys. The probability of small covered intrusive bodies within the claim area is high. The western third of the grid area is underlain by volcanics of largely latite composition. A quartz monzonite dyke was located during the 1998 programme.

Showings of massive sulphides and gold-bearing arsenopyrite in the Lemieux Creek and the Trench area northeast of Lemieux Creek appear fault associated though they could relate to more distal Besshi or Kuroko type deposits. The presence of pyrrotite enhances the applicability of ground magnetics.

b) PROPERTY GEOLOGY AND MINERALIZATION:

The paucity of outcrop exposure within the grid area requires utilisation of angular float as well as aerial photography to assist in the geological interpretation.

Much of the grid is covered by a layer of glacial till, probably less than 1.5 metres in thickness, laid down in the Pleistocene by a southerly moving ice sheet. The till layer is thickest at lower elevations, particularly around the shoreline and around the southern extension of Taweel Lake while in the upland sections it is thin or absent. Best outcrop exposures are around the cliffs and along the banks of many of the relatively steep draws.

Two principal rock types occur – a series of volcanic flows, sediments and porphyries of essentially latitic composition within the western third or quarter of the grid and a package of flysch sediments within the eastern two-thirds. A minor but significant exposure of quartz monzonite dyke rock trends north-northwesterly parallel to one of the fault structures associated with massive sulphide mineralization. GSC Memoir 363 assigns a Triassic age to the sediments, a Jurassic age to the volcanics while the dyke is probably age equivalent to the Baldy Batholith, ie – Cretaceous.

The contact between the volcanics and the flysch sediments trends north-northwesterly immediately east of Taweel Lake and is marked by a series of distinctive draws trending in that same direction. While the physical boundary between the two rock types is poorly exposed it would appear to be a fault contact.

The volcanics and the dyke are probably equivalent in terms of rock chemistry (ie - latite/monzonite) and may have a kindred association.

Massive sulphides are exposed in two locations – the Lemieux Creek adit (107e-116n) and the area of trenches (111e-121+50n). The material at Lemieux Creek consists of massive pyrite/pyrrotite with minor arsenopyrite and secondary scorodite in a breccia within flysch sediments. The massive sulphides in the trench area also occur within brecciated flysch-type sediments, with or without associated quartz and/or calcite veining, and contain iron-rich sphalerite, galena, chalcopyrite and pyrite. Both zones have a significant gold content (Jenks 1994).

In the field fault lineaments are seen to generally coincide with sharp, linear draws.

Table 1 - Descriptive Statistics Taweel Lake Soll Geochemistry Sep-02

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	Мо	Cu	Pb 2	Zn	Ag	Ni	Co I	Иn	Fe	As	U,	Au '	Th	Sr	Cd	Sb	Bi	с	Ca	P	La	Cr	Mg	Ba	TI	B	Al	Na	К	W H	lg s	ic Ti	S	Ga	ł
Number	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	50	1	50 50	5() 5	0 50	0 50	50	50	50	50	50	50
Minimum	0.5	2.4	4.9	18	0.1	1.1	0.8	26	0.31	1.3	0.1	0.5	0.4				0.1		0.05			3.4		-	24 0.016			-	4 0.02		0.01				3
Maximum	11.5	98.8	23.5				15.7		4.27	73.4	5.2	5.3	5.1	92	3.3	2.9	12	65	0.75	0.51	20	82.6	0.74	2	74 0.081	1	3 34	8 0.014	4 0.17	0.2	0.25	10.1	0.3	0.1	10
.		16.622		148.1	0.418	24.632	8.148	684.14	2.304	15.608	0.464	1.465	2.348	20.18	1.36	1.212	0.268	41.3	0.195	0 144	12 28	21 404	0.24	109	A 0.04F	1.29	8 1.40	1 0.00	7 0.059	0.104	0.037	2,032 0	.123 0	.089 (8.08
Geometric	2.505	13.914	12.839	136.051	0.338	21.063	7.494	449.484	2.177	12.393	0.362	1.248	2.12	17.61	1.185	1.014	0.229	40.208	0.169	0.11	11.096	19,989	0.07	95.7	27 0.043	1.22	2 1.28	9 0.00	7 0.056	0.103	0.032	1.840 U	0.117 U	1,001 3.	.0/1
Mode	2.0	18.2	12.1	165	0.4	18.8	6.2 1	n/a	1.98	11.8	0.4	0.7	1.4	14	0.6	0.5	02	40	0 13	0.093	11	18 1	0.22		67 0.044		1 1.1	4 0.00	6 0.05	0.1	0.04	1.8	0.1	0.1	0
Standard (1.998	13.691	3.638	55.719	0.35	15.401	2.977	729.406	0.69	12.255	0.695	0.974	0.97	13.301	0.733	0.733	0 212	9.033	0 110	0 110	5 725	8 283	0 453	50 A	08 0.014	0.50	7 0.54	3 0.00	2 0.024	0.02	0.033	1.278 0	.048 0	0.015 1.	.583
Skewness	2.275	9 4.5	0.525	0.42	3.329	3.656	0.362	2.582	0.223	2.585	6.701	2.139	0.54	3.461	1.028	0.954	3.624	-0.024	2.313	1.622	1.358	2.385	A 875	4 4	38 A 615	1 42	5 0 0 3	2 1 05	1 2.454	4.683	5.442	5.264	1.81 +1	. 44 0 U.	
		26.949		0.386	13.364	20.213	0.4513	8.51	1.294	9.325	46.442	5.7	0.87	16.921	0.506	0.105	14.143	1.531	8.576	2.243	2.142	11.939	0.3	1.6	24 0.11	1.1	3 3.32	8 2.16	1 9.639	20.819	34.62	33.001 3	0.031 1		0.21
Upper qua								783.75	2.703	1 8 .1	0.4	0.5	2.775	23	1.7	1.6	0.3		0.248		14	25.45	0.3775	120	5.5 0.05	j	2 1.72	3 0.00	8 0.068	0.1	0.04	2.275	0.1	Q. I	· · ·
Upper dec	1 5.04	28.21	17.22	225.5	0.61	36.06	11.92	1457.4	3.091	32.9	0.6	2.44	3.33	35	2.41	2.41	0.4	51	0.331	0.287		28.02			87 0.06		2 1.92	8 0.00	9 0.08	0.1	0.051	2.61	0.2	0.1	8

Table 2: TAWEEL LAKE SOIL SAMPLING CORRELATION COEFFICIENTS Sept 2002

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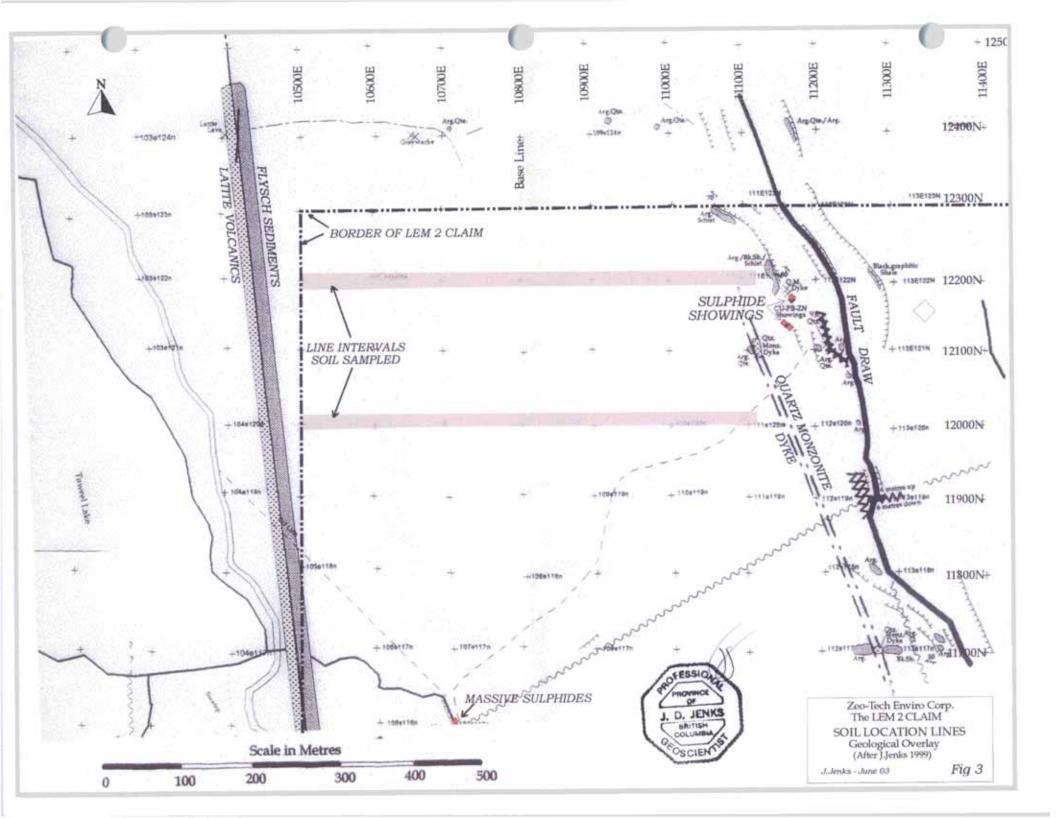
Correlat	tion Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	υ	Au	Th	Sr	Cd	Sb	Bi	v	Ca	P	La	Cr	Mg	Ba	Ti	в	Al	Na	к	W	Hg	Sc	TI	S	Ga
Мо		0.42	3 0.402	2 0.178	3 0.026	0.324	0.050	-0.118	0.547	0.416	0.273	-0.024	0.273	0.203	0.158	0.677	0.207	0.323	0.111	-0.280	0.333	0.288	0.254	-0.060	-0.397	-0.026	0.081	-0.205	0.243	-0.046	0.129	0.189	0.264	-0.223	-0.126
Сц	0.423	3	0.504	0.354	4 0.575	0.923	0.563	0.213	0.680	0.500	0.906	0.447	0.541	0.796	0.385	0.614	0.253	0.555	0.713	-0.213	0.613	0.860	0.601	0.349	-0.225	-0.122	0.612	0.312	0.730	-0.089	0.815	0.911	0.571	0. 445	0.179
Pb	0.402	2 0.504	ļ 1	0.406	6 0.305	0.473	0.421	0.093	0.67 8	0.580	0.475	0.369	0.553	0.479	0.340	0.872	0.504	0.494	0.401	0.090	0.484	0.570	0.373	0.307	-0.074	-0.003	0.541	0.249	0.556	0.160	0.374	0.503	0.375	-0.325	0.499
Zn	0.178	3 0.35	0.406	3 '	1 0.296	0.550	0.725	0.375	0.520	0.323	0.284	0.293	0.514	0.358	0. 549	0.273	0.231	0.092	0.315	0.429	0.314	0.471	0.385	0.689	0.094	0.172	0.632	0.251	0.584	-0.098	0.299	0.390	0.374	0.169	0.453
Ag	0.020	0.57	5 0.305	5 0.29	61																	0.534													
Ni	•··	-			0 0.564	•																0.887													
Ċo					5 0.409																	0.698													
Mn					5 0.462																	0.092													
Fe					0.259																	3 0.832													
As					3 0.015																	3 0.495													
U					4 0.689																	2 0.793 4 0.463													
Au					3 0.456																	0.679													
Th					4 0.096																	0.776													
Sr					8 0.632 9 0.443																	0.311													
Cd	0.15	3 0.38	0 0.340	0.04	9 0.443 3 0.065	0.313	0.400	0.007	0.240	0.155	0.417	0.363	0.104	0.346																					
Sb					1 -0.003													0.398	0.270	-0.123	0.021	0.340	0.580	-0.090	-0.128	-0.076	0.356	0.115	0.415	0.023	0.005	n.26	3 0.156	0.040	0,199
					2 0.207																	5 0.658													
v Ca					5 0.522																	0.711													
P	.02	0.71 R_0.21	3 0.40	0.31	9 0.131	0.071	0.000	0.219	-0.011	-0.253	-0.076	0.017	0.016	0.052	0.383	-0.367	-0.123	-0.303																	
La	0.2	3 0 61	3 0.00	4 0.31	4 0.132	0.644	0.531	-0.034	0.708	0.693	0.512	0.494	0.864	0.427	0.134	0.621	0.667	0.465	0.454	-0.296	1	0.670	0.824	0.057	-0.371	-0.036	0.496	0.059	0.645	-0.172	0.234	0.56	3 0.276	0.137	0.169
Cr	0.00	B 0.86	0 0.57	0.47	1 0.534	0.887	0.698	0.092	0.832	0.495	0.793	0.463	0.679	0.776	0.311	0.557	0.340	0.658	0.711	0.012	0.670) 1	0.743	0.355	0.050	0.016	0.820	0.310	0.767	' -0.019	0.666	0.89	3 0.380	0.507	0.504
Mg	0.25	4 0.60	1 0.37	3 0.38	5 0.091	0.648	0.668	-0.090	0.770	0.647	0.379	0.355	0.813	0.431	0.023	0,583	0.580	0.596	0.495	-0.203	0.824	0.743	1										1 0.145		
Ba					9 0.536																			1									3 0.345		
Ťì	-0.4	0 -0.22	5 0.07	4 0.09	4 0.104	-0.115	0.103	-0.069	-0.029	-0.325	-0.093	-0.097	-0.091	0.033	-0.028	-0.314	-0.128	0.119	0.002	0.515	-0.371	1 0.0 50	-0.085	0.055									5 -0.300		
B	-0.0	3 -0.12	2 -0.00	3 0.17	2 0.002	-0.056	0.084	0.231	0.040	-0.116	-0.057	-0.034	-0.064	0.121	0.319	-0.177	0.076	-0.088	0.15	0.179	-0.036	3 0.0 16	-0.045	0.168	0.134					-0,138			2 0.077		
AJ	0.08	1 0.61	2 0.54	1 0.63	2 0.529	0.751	0.797	0,111	0.729	0.392	0.642	0.520	0.703	0.618	0.3 86	0.337	0.356	0.414	0.543	0.363	0.496	3 0.8 20	0.590	0.462	0.312	0.112	1	0.430	0.683	0.088	0.567	0.80	J 0.279	0.309	0.730
Na	-0.2	1 0.31	2 0.24	9 0.25	1 0.508	0.333	0.294	0.230	0.035	0.032	0.516	0.532	0.118	0.456	0.331	-0.081	0.115	0.011	0.345	0.312	0.059	9 0.30 9	-0.058	0.417	0.252	-0.05	0.430					-	0 0.321		
К	0.24	3 0.73	0 0.55	6 0.58	4 0.482	0.794	0.668	0.282	0.642	0.457	0.740	0.490	0.668	0.761	0.458	0.416	0.415	0.459	0.769	0.030	0.645	5 0.767	0.571	0.530	-0.092	0.015	0.683	0.428	1			• · · · ·	3 0.495		
W	-0.0	5 -0.08	9 0.16	0 -0.09	8 0.070	-0.116	-0.055	5 -0.041	0.079	-0.073	-0.054	-0.125	-0.030	-0.135	0.163	0.129	0.023	0.170	-0.124	0.046	-0.172	2 -0.019	-0.049	-0.114	0.236	-0.136	0.088	-0.017	-0.100) 1			7 -0.110		
Hg	0.12	9 0.81	5 0.37	4 0.29	9 0.736	0.752	0.439	0.404	0.355	0.119	0.899	0.357	0.222	0.818	0.515	0.219	0.005	0.344	0.744	0.067	0.234	4 0.666	0.220	0.531	0.081	-0				0.039	-		9 0.550		
Sc	0.18	9 0.91	1 0.50	3 0.39	0 0.651	0.902	0.627	0.203	0.641	0.373	0.949	0.504	0.590	0.825	0.389	0.423	0.263	0.524	0.746	0.018	0.566	B 0.893	0.561	0.433	0.045	-0.052	0.800	0.480	0.783	0.017	0.869	; 	1 0.504		
TI	0.26	4 0.57	1 0.37	5 0.37	4 0.460	0.512	0.296	0.295	0.347	0.377	0.575	0.316	0.204	0.515	0.339	0.365	0.156	0.324	0.380	-0.206	0.276	5 0.380	0.145	0.345	-0.300		0.279	0.321	0.495	0.0110	0.000	v v.50	-) 0.057 I -0.351
S	-0.2	2 0.44	5 -0.32	5 0.16	9 0.304	0.383	0.594	0.296	0.181	0.152	0.265	0.065	-0.006	0.778	0.544	-0.378	0.040	0.112	0.919	-0.235	0.137	/ 0.507	0.507	0.391	-0.092	0.344	0.309	-0.141	0.565	0.000	0.000	> U./O > n.∡o		/ 1 7 n 3#/	-
Ge	-0.1	3 0.17	9 0.49	9 0.45	3 0.333	0.336	5 0.430	0.021	0.498	0.093	0.307	0.211	0.428	0.332	0.209	0.120	0.199	0.328	0.272	0.523	0.168	9 0.504	0.225	U.343	0.581	0.211	0.730	0.209	0.423	5 0.000	0.209	• Q.43	+ 0.001	-0.301	I 1

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5. GEOCHEMICAL SOIL SURVEY 2002:

a) GENERAL:

Four days were spent by the author on the property between September 19th through the 22nd, 2002. Portions of the period were lost due to heavy rainfall and time spent negotiating access to the claim area through the Nehalliston Lake fishing camp.

In total fifty soil samples were collected with a mattoch at twenty-five metre spacings along two different pre-cut grid lines located 200 metres apart. This represents a lateral interval totaling 1,250 metres. While sample depths ranged from 15 to 20 centimetres the usual depth was 15 cm. Target material was "B" horizon, however, juvenile soil development at many sites produced a sample which would be classed as weathered "C". The samples were placed in kraft waterproof sample envelopes and submitted to Acme Analytical Laboratories Ltd. in Vancouver for analysis. In places where they had deteriorated lines were re-numbered and re-flagged during sample collection.

Information recorded included vegetation type, sample depth, soil characteristics, rock type, drainage direction and degree of slope (see Appendix 1).

Samples were under the author's custody at all times and personally submitted to the laboratory after the survey was completed.

b) SOIL TYPES:

It was found that while soil profiles varied somewhat they fell within two principal categories:

a) Podzols: Typically these consisted of ashy soils with slightly to poorly-developed soil horizons. They comprised an upper organic layer anywhere from 4 – 10 centimetres thick followed by a light-grey ashy layer succeeded by a dark to reddish-brown so-called "B" horizon underlain by weathered rock fragments making up the "C" horizons. In places where swampy conditions prevailed the organic layer was extremely deep and suitable sample material was difficult to obtain.

b) Azonal Juvenal Soils: Along steeper slopes and in areas of high outcrop.percentages soils tended to be poorly developed consisting of partially weathered rock debris reflective of the underlying rock type. The eastern (east of 110E) portions of the sample lines tended to fall within this category.

In certain areas the two varieties were mixed to a degree. In at least one location (120N-105+50E) the material was extremely clay-rich and appeared to be glacio-lacustrine in origin.

c) ANALYTICAL METHODS:

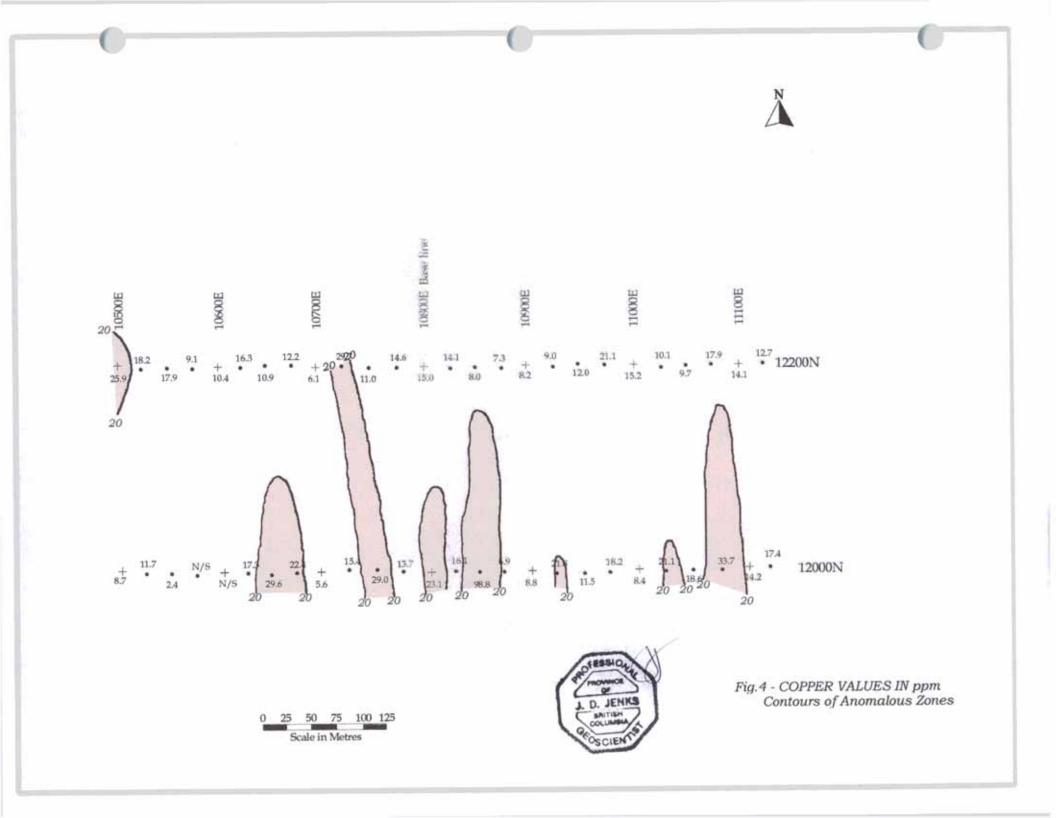
The samples were submitted to Acme Analytical Laboratories in Vancouver where they were analysed as per Acme's 1DA package consisting of a 35 element determination (Appendix II) by Inductively Coupled Plasma (ICP) and Mass Spectrography (MS). Ten gram sample portions are leached in hot aqua regia for one hour and subjected to ICP/MS analyses. Detection limits for gold are 0.5 ppb. Detection limits for all the elements are shown in Appendix II as well as analytical results.

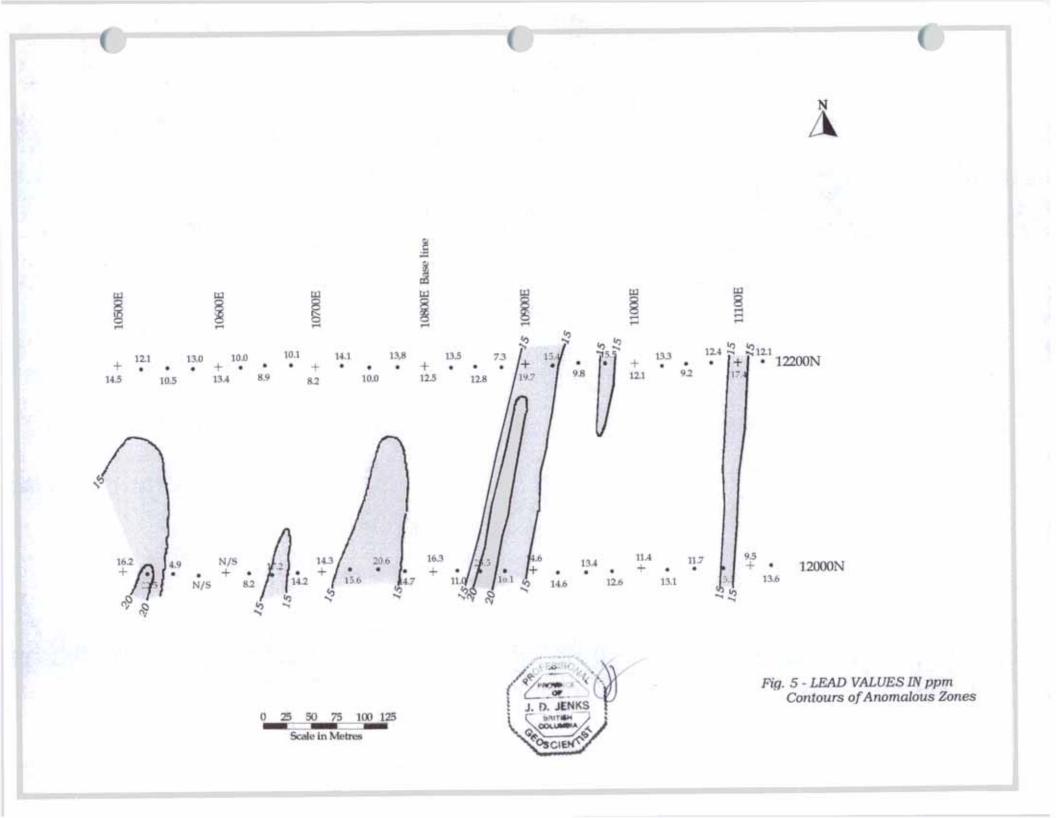
d) DESCRIPTIVE STATISTICS:

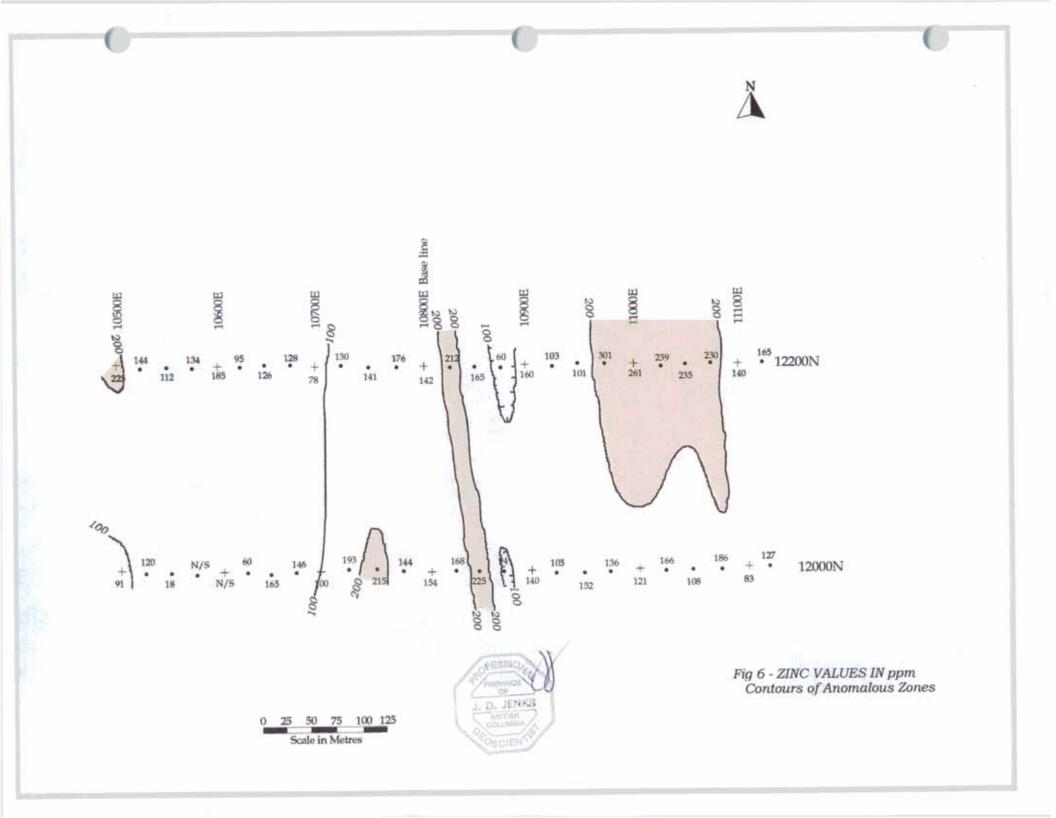
Table 1 summarizes the various statistical parameters for the thirty-five elements analysed as part of Acme's 1DA package. These parameters focus upon measures of central tendency (arithmetic mean, geometric mean, mode), distribution of values (skewness, kurtosis) and variation of values (standard deviation) of the sample population from the norm. Exactly what may be regarded as anomalous generally

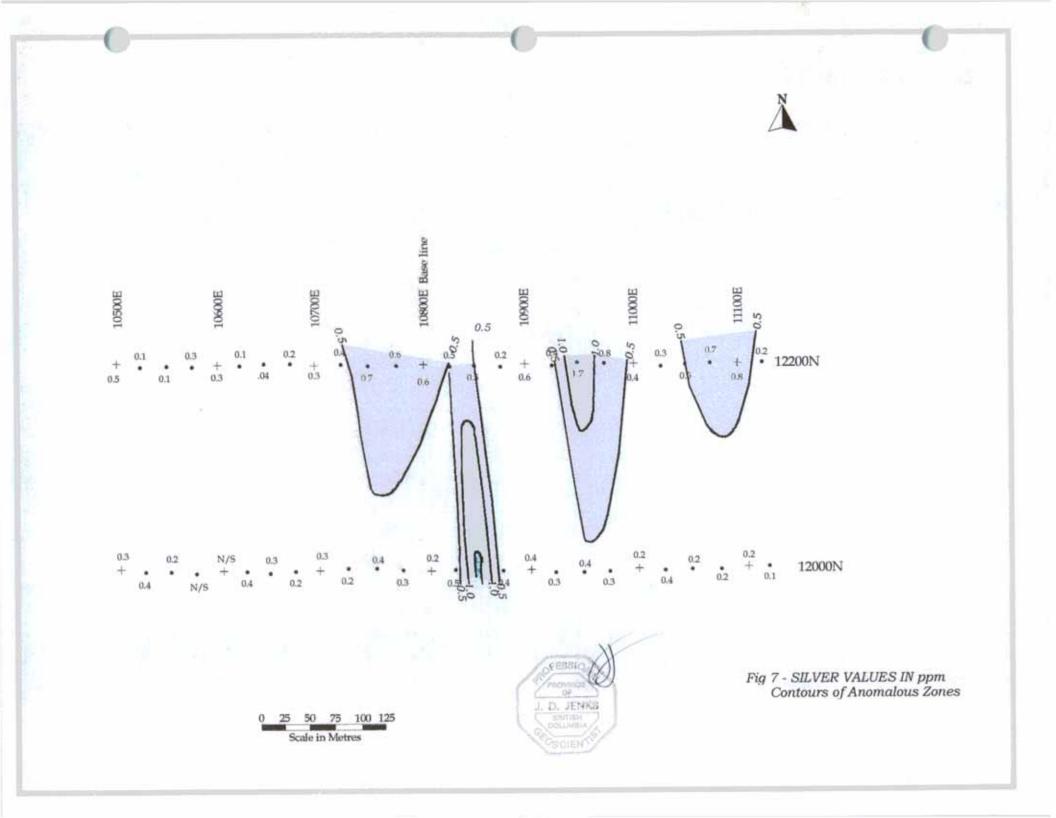
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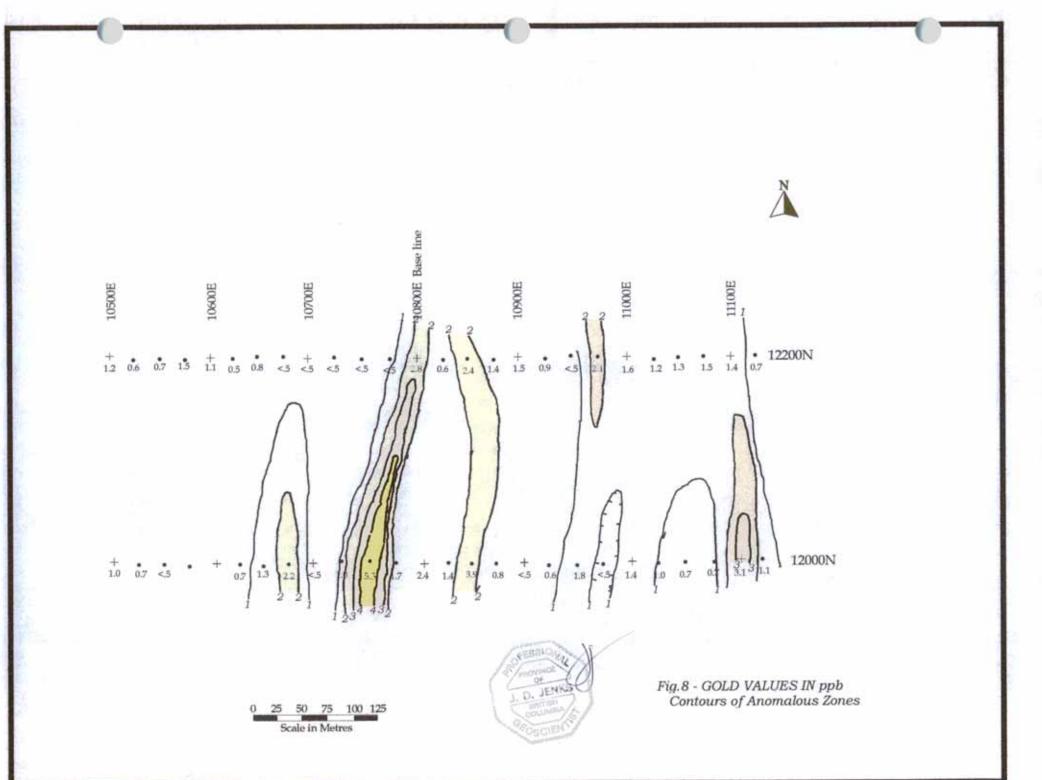
i

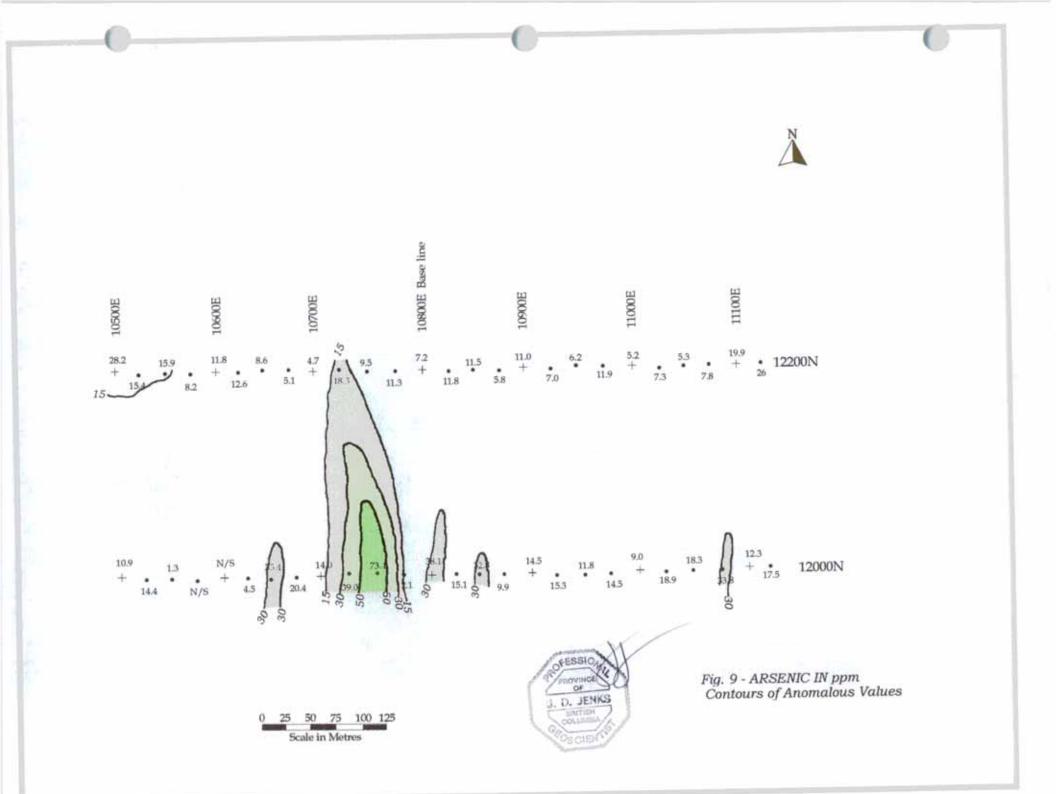












involves consideration of arithmetic mean and standard deviation (X + 2SD) and/or values of upper quartile and decile.

Correlation coefficients are listed in Table 2. They reflect the relationship of a given element to another. In other words a perfect correlation between two elements is reflected as 1. Gold is one element in particular which, due to its dispersion and lab analysis characteristics, may be better reflected by one or more of its various pathfinder elements such as As, Ab or Bi. Arsenic anomalies are shown in Figure 9.

The main elements of focus on the property are copper, lead, zinc, silver and gold. Arsenic with a correlation coefficient 0.512 for gold is considered along with the main elements. Anomalies for these various elements are well indicated by their upper quartile and decile values. Accordingly anomalous distributions for these elements are presented in Figures 4 through 9.

6. RESULTS AND DISCUSSION:

Sample analytical results are presented in Appendix II. Results and locations are also listed in Figures 4 through 9 for the elements Cu, Pb, Zn, Ag, Au and As as well as depiction of anomalies.

Due to the relatively small number of samples taken anomalies are often predicated on the values of 1-3 samples. Accordingly, they should be viewed in that context and with some caution. Anomaly compilations are illustrated in Figures 10 and 11.

Essentially three anomalous zones emerge from the survey designated A-A1, B-B1 and C-C1 ranked in decreasing order of significance.

Anomaly A-A1:

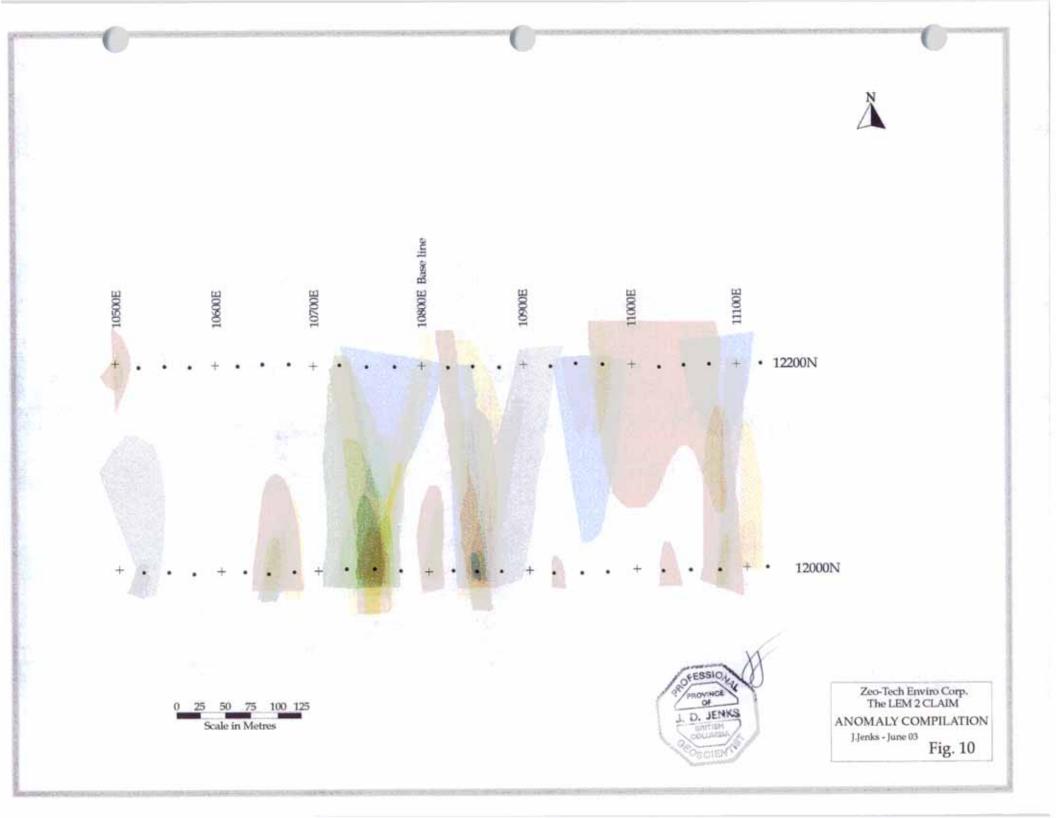
A north-northwesterly somewhat linear anomaly A-A1 trends parallel to the regional structure. No outcrop is exposed within the zone, however, the underlying bedrock is a flysch/greywacke/argillite. This particular anomaly has strong elevated coincidental values of gold, copper, zinc and silver all trending in the same direction

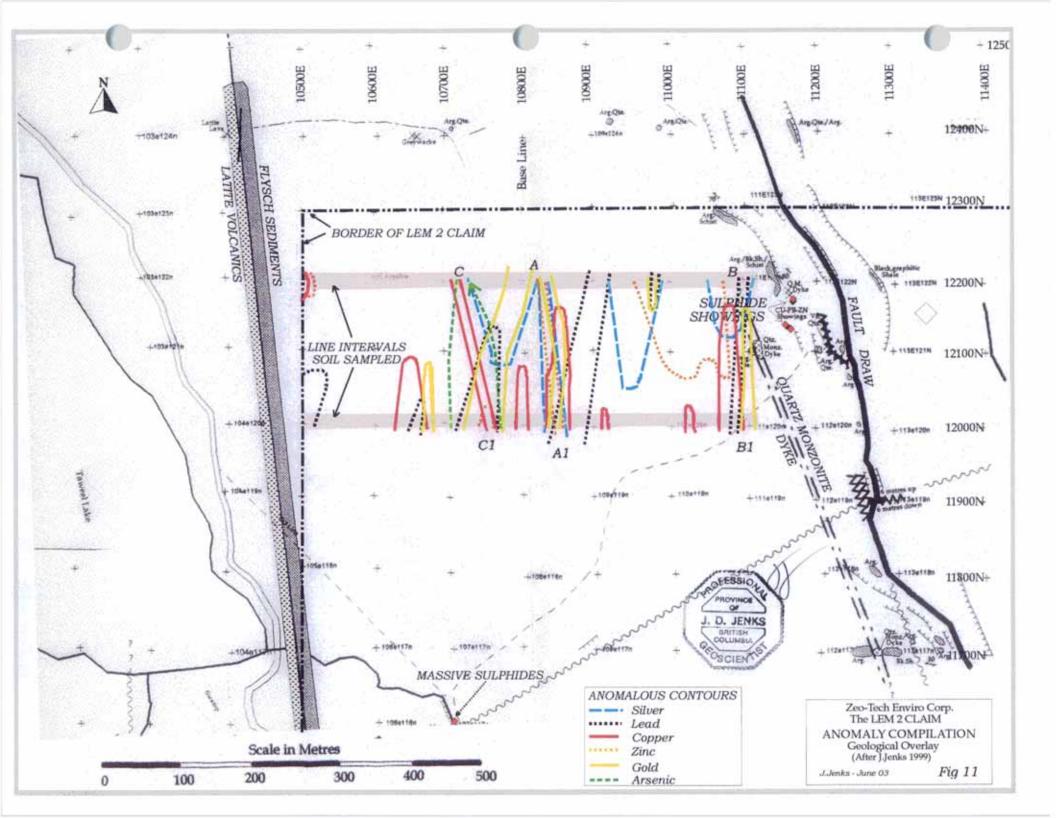
Anomaly B-B1:

A northerly-trending anomaly with coincidental high lead, copper, silver, gold and zinc values. It is located fifty metres west of one of the principal Cu-Pb-Zn sulphide showings on the property within an area of fairly abundant outcrop – greywacke and argillite. In addition a mapped quartz-monzonite dyke projects onto the northern portion of the anomaly. A major NNW-trending fault draw is situated 125 metres east of anomaly B-B1.

Anomaly C-C1:

Apparently running parallel to anomaly A-A1 some 100 metres to the west. Anomaly C-C1 is also within an area of no outcrop. It is made up of high coincidental values of copper, arsenic, silver and to some degree, gold, in the southern portion. There is a suggestion of a possible cross-structure as elevated gold, silver and lead values trend obliquely to the prime anomaly.





7. CONCLUSIONS:

The survey delineated three anomalous areas with significant potential. Anomalies A and C are located in areas of no outcrop. As such they could represent some sort of surface manifestation of mineralization that has not yet been exposed. Anomaly B lies adjacent to a major fault structure, known mineralization and is in an area through which a monzonite dyke projects.

While preliminary results are encouraging it would be in order to extend the survey to the south and across the fault draw to the east. Some duplicate sampling of anomalous areas would be in order for verification purposes.

8. RECOMMENDATIONS:

- Extend the survey to the south and east at the sample spacings undertaken to date.
- Prospecting in and around the vicinity of anomaly B-B1.
- Increase sample density within anomalous areas in an effort to quantify anomaly strength and better define target locations.
- Survey the lines with a low-cost type of electromagnetic survey such as an EM-16 in an attempt to establish whether coincidental conductors are present beneath the anomalous zones.

Much of the above could be completed for \$10,000 - \$15,000.

9. STATEMENT OF EXPENDITURE:

The following expenditures were incurred on the Lem 2 claim during the anniversary year ending March 25, 2003:

Consulting fees including sample collection, travel, one rain day		\$950.00
Interpretation of data, report preparation		500.00
Accomodation		221.95
Meals		114.28
Vehicle expense:		
Gas	\$77.00	
Rental & kilometrage	312.80	389.80
Sampling supplies and equipment		77.81
Laboratory charges		<u>695.50</u>

Total

\$2,949.34

Respectfully submitted,

John Jenks - P.Geo. (B.C.) June 23, 2003



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CERTIFICATE OF AUTHOR

I, John Jenks, do hereby certify that:

- 1. I am an independent geological consultant resident at the above address.
- I graduated with a B.Sc. (Geology Major) degree from McGill University Montreal, Canada in 1968. 2.
- 3. 1 am a Registered Professional Geoscientist (#21123) with the Association of Professional Engineers and Geoscientists of British Columbia as well as a Life Member of the Association of Professional Engineers, Geologists and Geophysicists of Alberta.
- 4. I have worked as a geologist for a total of thirty-five years since my graduation from university.
- 5. I have read the definition of "qualified person" set out in National Instrument 43-101 ("NI 43-101)) and certify that by reason of my education, affiliation with a professional association (as defined in NI 43-101) and past relevant work experience, I fulfill the requirements to be a "qualified person" for the purposes of NI 43-101.

6. I am responsible for the preparation of the technical report entitled "Geochemical Soil Survey - the Lem 2 Mineral Claim" and dated June, 2003 relating to the Lem 2 mineral claim, Kamloops Mining Division, BC. I have visited the property many times since my initial visit in the early 1990's and have carried out various work activities including mapping and rock sampling. The date of my last visit was September 22, 2003.

Dated this 22 Day of June, 2003

Signed



Appendix I:

J.JENKS - FIELD NOTES

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J.Jenks - Field Notes Lem 2 Soll Sampling Sep-02 In Account with Zeo-Tech Enviro Corp.

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Sample #	Grid Station	Vegetation Type	Sample Depth (Centimetres)	Sample Description	Drainage Dir.& Slope	Other
2LS -	122N -					
1	105E	Thick conifers - spruce, bals.fir	10	Lt brwn, rocky	240 deg @ -25 deg	
2	105+25E	85	10	Lt.brown	260 d @ -25 d	Greywacke frags.
3	107 +50 E	łı	12	Grey brn, ashy pod.	40	
· 4	105+75E	16	15	Ashy red	260 d @ -15 d	Greywacke frags.
5	106E	li i	12	Lt.grey-brwn,ashy	260 d @ -20 d	
6	106+25E	Balsalm fir, poplar, spruce	12	Lt.grey - rusty brwn	220 d @ -20 d	-
7	106+50E	Balsaim fir, spruce	15	Ashy grey	220 d @ -15 d	
8	106+75E		15	17	67	
9	107E	te	15	Dark brown	220 d @ -20d	Rocky,greywacke
10	107+25E	£8	15	Brown, ashy	180 d @ -10 d	Greywacke
11	107+50E	60	15	H	180 d @ -5 d	5 cm org,3cm ashy
12	107+75E	99	15	Dk.brown	200 d @ -5 d	
13	108E (BL)				-	
	120N -					
14	108E (BL)	Balsalm fir, spruce	15	Lt.grey brown,ashy	180 d @ -20 d	
15	107+75E	0	15	11	180 d @ -30 d	
16	107+50E	Balsalm fir, spruce, pine	15	Lt.reddish brown	210 d @ -25 d	107+40E: small draw @180 d
17	107+25E	e9	15	Lt.redd.brown,ashy	180 d @ 30 d	
18	107E	••	15	5 cm org,grey ashy	210 d @ -15 d	Reddish brown
19	106+75E	64	15	dk brw, some clay	210 d @ -5 d	
20	106+50E	Spruce, balsalm fir	30	org,dk brn,some red	**	
21	106+25E	11	20	Brown, ashy	180 d @ 3 d	
No Sample	106E	91		Deep organic layer	-	
No Sample		u		H	190 d @-2 d	
	105+50E	u	20	Org,silt, clay	045 d @ -2 d	Glacial lacustrine
	105+25E	u		Dk brw, stoney	240 d @ -3 d	

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24 105E	U	15 Lt.grey ashy,	270 d @ -5 d	Reddish brown beneath
122N-				
25 108+25E	Spuce, balsalm fir, pine	15 Lt to redd brn, ashy	200 d @ -10 d	
26 108+50E	1 1	20 "	210 d @ -10 d	Upper 6 cm is ashy
27 108+75E	u	20 Lt rusty brown	210 d @ -20 d	Quartzite/greywacke
28 109E	14	20 Lt redd brn,silty loam	210 d @ -5 d	48
29 109+25E	N	20 "	17	17
30 109+50E	H	15 Dark brown, talus	170 d @ -10 d	Rusty black shale
31 109+75E	14	20 Org,ashy grey-rd brw	200 d @ -15 d	Arg qte with pyrite
32 110E	54	20 Lt grey,ashy,rocky	17	Greywacke
33 110+25E	н	20 Lt grey to redd brn	**	H
34 110+50E	11	15 Light reddish brown	200 d @ -20 d	H.
35 110+75E		15 Lt brown rocky soil	160 d @ -15 d	-
36 111E	Spr, bal fir, pine, cedar	15 Rdd brn rocky thin soll	130 d @ 20 d	Argillaceous quartzite
37 111+25E	10	15 Thin, dk brn, rocky	140 d @ -35 d	
120N-				
38 111+25E	Thick cover spruce, balsalm fir	15 Lt.brown, rocky	310 d @ -10 d	Rocky outcrop
39 111E	u	15 Lt.brown, loamy	160 d @ -5 d	Charcoal layer, trenches
40 110+75E	18	15 "	м	
41 110+50E	LF	15 "	N	Greywacke, arg quartzite
42 110+25E	te	15 "	160 d @ -10 d	
43 110E	61	15 "	H.	Greywacke, quartzite
44 109+75E	61	15 "	n	41
45 109+50E	44	20 Reddish brown, loamy	175 d @ -15 d	Greywacke
47 109+25E	(I)	20 Dark & reddish brown	n	
46 109E	Balsaim fir, spruce, poplar	20 Org 8 cm, It gr ashy, r.brn	47	
48 108+75E	H.	20 Light brown, ashy	210 d @ -15 d	
49 108+50E	i.	20 Dk to redd brn, clay rich	200 d @ -20 d	
50 108+25E	10	20 Light reddish brown	185 d @ -15 d	

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Appendix II:

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ANALYTICAL TECHNIQUES AND RESULTS

GEOCHEMICAL - ICP by Aqua Regia Digestion

GROUP 1C MERCURY BY COLD VAPOUR AA OR ICP-MS

Accurate, low level determination of Hg by Aqua Regia digestion followed by either cold vapour AA or ICP-MS analysis.

Element	Method	Detection	Cdn	U.S.
Hg	Cold Vapour AA or ICP-MS	10 ppb	\$4.40	\$3.30
Hg	Cetac Cold Vapour AA	1 ppb	\$7.70	\$5.80

Analysis is not suitable for high-grade Au, Pt or elevated Se samples (cold vapour method only). Acme retains the right to select the method of determination.

GROUP 1D, 1DX: ICP & ICP-MS ANALYSIS - AQUA REGIA

You can choose economically priced ICP-ES (Group 1D) or ICP-MS (Group 1DX) analysis to complement your exploration program. Sample splits of 0.5 g are leached in hot (95°C) Aqua Regia. Select a larger split size for more representative Au analysis. Au results may not be quantitative in massive sulphide and graphitic samples.

Group 1D	Cdn	U.S.
Any 1 element	\$3.85	\$2.90
Any 5 elements	\$5.20	\$3.90
All 30 elements	\$6.35	\$4.75
\$Include Hg and Tl add	\$0.50	\$0.40

Group 1DX

Any 1 element	\$7.00	\$5.25
Any 5 elements	\$8.00	\$6.00
All 36 elements	\$9.00	\$6.75
15 g sample add	\$3.50	\$2.65
30 g sample add	\$5.00	\$3.75

See Page 6 for Group 1F-MS Aqua Regia / ICP Mass Spec analysis for ultratrace element determination

	Group 1D Detection	Group 1DX Detection	Upper Limit
Ag	0.3 ppm	0.1 ppm	100 ppm
	0.01 %	0.01 %	10 %
As	2 ppm	0.5 ppm	10000 ppm
Au	2 ppm	0.5 ppb	100 ppm
B*	3 ppm	1 ppm	2000 ppm
Ba*	1 ppm	1 ppm	1000 ppm
Bi	3 ppm	0.1 ppm	2000 ppm
Ca*	0.01 %	0.01 %	40 %
Cd	0.5 ppm	0.1 ppm	2000 ppm
Co	1 ppm	0.1 ppm	2000 ppm
Cr*	1 ppm	1 ppm	10000 ppm
Cu	1 ppm	0.1 ppm	10000 ppm
Fe*	0.01 %	0.01 %	40 %
Ga*	-	1 ppm	1000 ppm
Hg‡	1 ppm	0.01 ppm	100 ppm
K*	0.01 %	0.01 %	10 %
La*	1 ppm	1 ppm	10000 ppm
Mg*	0.01 %	0.01 %	30 %
Mn*	2 ppm	1 ppm	10000 ppm
Мо	1 ppm	0.1 ppm	2000 ppm
Na*	0.01 %	0.001 %	10 %
Ni	1 ppm	0.1 ppm	10000 ppm
P*	0.001 %	0.001 %	5 %
Pb	3 ppm	0.1 ppm	10000 ppm
S		0.05 %	10 %
Sb	3 ppm	0.1 ppm	2000 ppm
Sc		0.1 ppm	100 ppm
Se		0.5 ppm	1000 ppm
Sr*	1 ppm	1 ppm	10000 ppm
Th*	2 ppm	0.1 ppm	2000 ppm
Ti*	0.01 %	0.001 %	10 %
4 T	5 ppm	0.1 ppm	1000 ppm
U*	8 ppm	0.1 ppm	2000 ppm
۷*	1 ppm	2 ppm	10000 ppm
W*	2 ppm	0.1 ppm	100 ppm
Zn	1 ppm	1 ppm	10000 ppm

*Some elements are partially leached.

								Zeo	<u>-Te</u>	ch	Env	rird	5 C	ori	<u>.</u>	F	11e	#	A2	1FI 042 c v6e	57	Į	ag												
MPLE#	Мо ррпл	Cu ppm	Pib ppm	Zn ppm	Ag ppm	и ррп	Co Ippm	Mn ppm	Fe %	As ppr	ປ ippm	Au ppb	Th ppm	Sr ppm	Cd ppm	Sb ppm	Bi ppm	v ppm	Ca %		La ppm	Cr ppm											тi ppm		
.S-2 e-3	9.2	25.9 18.2 17.9	14.5 12.1 10.5	225 144 112	.5 .1	30.9 24.3 24.7	3.8 6.2 7.4 7.0 7.5	230 280 279	3.77 2.29 2.23	28.2	.6	1.2 .6 .7	2.3	24 14 11	1.6 1.3 .5	2.8 1.4 2.0	.2 .2 .2	50 43 40	.09 .16 .14	.093 .137 .060	11 11 14	24.9 19.1 18.1	.32 .34 .37	106 74 67	.035 .033 .033	2 1 1	1.46	.005	.08 .04 .04	.1 .1 .1	.04 .03 .02	1.7 1.8	.3<. .2<. .1 . .1<.	05 08 05	5
5-6	2.7	16.3 10.9	10.0 8.9 10.1	95 126 128	.1 .4	27.3	8.3 6.6 6.3 6.1 5 4.5	205 426 2488	2.32	12.6 8.6 5.1	5.3 5.3 5.2	.5 8. 2.>	2.7	17 19 11	.8 2.3 2.9	1.2 .5	.2	41 33 28	.19 .26 .10	.064 .131 .160	14 10 7	20.8 15.6 12.6	.41 .26 .13	67 101 182	.046 .042 .029	1 2 <1	1.23	.005 .006 .006	.04 .05 .04	.1 .1 .1	.02 .05 .04	1.9 1.8 1.3	.1<. .1<. .1 .1<. .1<. <.1<.	05 10 05	•
LS-10 LS-11 LS-12 LS-13 LS-14	1.3	11.0 14.6 15.0	10.0 13.8 12.5	141 176 142	.7 .6 .6	20.0 22.0 31.4) 10.6 9.4) 10.1 6 8.4 9 11.9	554 1409 247	2.13	9. 11. 7.	5.3 5.4 2.4	<.5 <.5 2.8	2.1 1.4 2.5	15 17 12	.6 1.0 1.3	.5 .6 .8	.2	40 44 37	.18 .14 .11	.218 .226 .195	7 9 10	20.3 25.3 23.6	.34 .38 .32	81 169 72	.059 .061 .061	1 2 1	1.80 1.77 1.74	.007	.05 .05 .05	.2 .1 .1	.06 .04 .04	2.3 2.4 2.0	.1<. .1<. .2<. .1<.	.05 .05 .05	
LS-15 LS-16 LS-17 LS-18 LS-19	4.7	29.0 15.4	20.6	215 193	.4	30.1	2 8.5 12.1 5 8.2 7 4.2 5 11.4	360 273 139	3.55 2.78 2.31	73.4 39.0 14.0	.6 .3 .3	5.3 1.0 <.5	4.4 3.2 2.2	40 18 11	2.5	2.9	1.2	51 49 49	.36 .21 .11	.061 .131 .230	26 15 12	26.8 25.5 16.3	.56 .44 .15	79 83 83	.020 .044 .044	1	1.81 1.47 1.34	.008	.09 .08 .06	.1 .1 .1	.03 .02 .03	2.5 2.0 1.6	.1<. .2<. .1 .1<. .1<.	.05 .09 .05	
LS-20 LS-21 LS-22 LS-23 LS-24	1.1	17.3 2.4 11.7	8.2 4.9 22.5	9 60 9 18 9 120	.4 .2 .4	10.1 1.1 13.0	5 12.4 6.2 .8 5 6.0 2 3.7	149 26 588	1.73	4.9 1.3 14.0	5.1 5.1 4.3	7. 4.5 7.	2.4 5.4 72.5	41 6 10	.6 .3 1.1	.4 5. 2.9	, .1 , .1	63 14 57	.39 .05 .09	.036 .010 .136	2 5 9	22.0 3.4	.32 .02 .24	48 24 82	.081 .028 .063	1 1	.69 19 1.53	.007 .009 .007	.05 .02 .05	.1 <.1 .2	.07 .01 .03	1.9 .5 1.7	.1. <.1. <.1<. .1<. .1<.	. 10 . 05 . 05	1
E 2LS-26 LS-25 LS-26 LS-27 LS-28	2.7	14.1 8.0 7.3	13.5 12.8 7.3	212 5 165 5 60	.5 .5 .2	26.9 15.0 12.9	3 11.3 5 10.8 5 11.8 5 5.7 5 8.1	1261 599 513	2.66	11.1 11.1 5.1	3.4 5.5 3.2	ء. 2.4 1.4	5 2.6 5 2.4 5 1.4	35 24 8	2.1 2.4 .6	.9 .4 .5	.2	34 35 41	.41 .16 .07	.450 .510 .049	9 6 8	21.1 26.6 21.4 13.1 16.1	.30 .17 .14	239 119 66	.056 .078 .041	1 2 1	1.81 2.39 .60	.008	.08 .05 .04	1. 1. <.1	.04 .04 .02	2.2 2.7 1.1	.1<. .1<. .1<. .1<.	.05 .05 .05	
LS-29 LS-30 LS-31 LS-32 LS-33	1.7 2.6 3.0	12.0 21.1 15.2	9.8 15.5 12.1	101 301 261	1.7 .8 .4	15.8 43. 32.	7 6.7 3 8.3 1 15.7 1 8.9 3 10.0	1893 1172 742	1.76	6. 11.9 5.2	2 .4 7 .4 2 .3	<.5 2.1 1.6	5 1.1 3.1 5 2.6	22 19 17	1.7 2.0 .9	.5 1.3 1.0	.2	38 37 35	.20 .14 .16	.104 .391 .092	11 12 14	19.4 27.5 21.0	.23 .35 .36	133 274 132	.045 .046 .042	1	1.19 2.10 1.35	.008 .008 .007	.04 .08 .07	.1 .1 .1	.03 .04 .02	1.4 2.4 1.8	.1<.	. 05 . 05 . 05	
TANDARD DS4	6.7	131.4	28.9	159	.3	33.	7 11.9	793	3.12	22.0	3 6.1	26.0	3.7	29	5.3	4.8	5.0	72	.55	.093	17	162.0	.57	137	.084	2	1.73	.033	.16	4.1	.27	3.7	1.1<,	. 05	(
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AA(<u> </u>										-																			(C	Δ	Δ
ACHE ANALYTICAL								Zeo	-Teo	ch i	Env	iro	C	or		E	'IL	E {	# A	204	25	7						Pa	ge	2		. –	ACHE AN	
SAMPLE#	Mo ppn	Cu ppm		Zn ippm	-	Ni ppm			Fe %	As ppm	U PPM	Au ppb	Th ppm	Sr ppm	Cđ ppm	sb ppm	Bi ppm	V mqc	Ca X		La ppm	Cr ppm		Ba ppm	Ti %	8 ppm	Al X	Na %	K X	W ppm	Hg ppm	Sc ppm		S Ga % ppm
G-1	.8	2.3	1.9	48	<.1	4.0	4.0	540	1.78	<.5	2.4	.7					.1	42	.46	.089		10.0				-							.3<.0	
2LS-34	1.6	9.7	9.2	235	.5		9.3	704	1.72	5.3	.4	1.3	2.2	14	1.2	.5				.281		15.2				-		.009				1.8	.2<.0	-
2LS-35	2.5	17.9	12.4	230	.7	27.1	9.1	4023	1.78	7.8	.3	1.5	1.5	20	3.2	.9				.084	11	16.5				_		.007				1.7	.2<.0	-
2LS-36	2.8	14.1	17.4	140	.6	16.5	8.0	1033	2.07	19.9	.4	1.4	2.3	23	1.8	1.0				.091	11	14.6						.006				1.6	.2<.0 .2<.0	
2LS-37	1.9	12.7	12.1	165	.2	27.9	11.5	453	2.22	26.0	.3	.7	3.0	15	.7	1.5	.3	40	.13	.037	14	16.6	.28	125	.044	1	1.37	.007	.07	• 1	.02	1.8	.25.0	ז כו
2LS-38	4.0	17.4	13.6	5 127	.1	25.2	7.0	709	2.40	17.5	.4	1.1	2.7	16	1.0	1.9	.2	43	.20	.070	18	19.0	.38	90	.040	1							.1<.0	
2LS-39	1.7	14.2	9.5	5 83		21.8		129	1.67	12.3	.3	3.1	1.4	10	1.0	.7	.2	44	.08	.020	-8	15.0	. 16	- 44	,040	1		.009					.1<.0	-
2LS-40	5.0	33.7	15.5	5 186	.2	38.6	9.7	362	3.10	33.8	.6			20						.045		27.0			.030			.007				2.3	.2<.0	
2LS-41	3.4	18.6	11.7	7 108		28.6								14						.046					.036			.005			.02		.1<.0	
2LS-42	3.7	21.1	13.1	166	.4	40.9	12.7	361	2.96	18.9	-4	1.0	2.7	31	1.1	1.5	.3	44	.33	.055	16	28.4	.49	78	.044	3	1.92	.006	.08	•1	.04	2.0	.1<.0	15 (
2LS-43	1.8	8.4	11.4	÷ 121	.2	14.9	6.4	1186	1.71	9.0	.3	1.4	2.1	14	1.2	.5	.2	31	.14	.126		19.8						.008					.1<.0	
2LS-44	3.3	18.2	12.0	5 136		28.9					.4	<.5	2.9	- 14	.6	1.3	.2	42	. 16	.098	16	28.0			.031			.006				2.0		-
RE 2LS-44	3.5	18.0	13.0) <mark>13</mark> 4	.3	28.1	8.1	372	2.38	14.1	.4	2.1	3.0	15	.6	1.3	.2			.099		27.3						.006				2.0	.1<.0	
2LS-45	2.6	11.5	13.4	152		22.1		216	2.80	11.8	.4	1.8	2.7	35	1.9	.9	.3			.334								.006				2.1	.1<.0	
2LS-46	2.4	8.8	14.6	5 140	.4	18.3	5.5	157	2.59	14.5	.3	<.5	2.3	14	1.2	.8	.3	45	.13	.203	13	22.1	.28	70	.050	Z	1.31	.005	.05	.1	.02	1.6	.1<.0	8 כו
2LS-47	11.5	21.8	14.6	5 105	.3	16.8	5.0	589	2.30	15.3	.4	.6	1.4	17	1.9	1.6	.2	37	. 15	.090	9	14.8	.15	70	.028	1		.006					.1<.0	
2LS-48	5.4	6.9	16.1	94	.4	9.9	3.8	206	1.89	9.9	.3	.8	1.1	13	.8	1.2	.2	40	.10	.057	9	15.9	.11	93	.016			.007				1.0	.2<.0	
2LS-49	5.7	98.8	23.5	5 225	2.1	111.2	15.5	1945	4.27	32.8	5.2	3.9	4.6	92	3.3	2.9	_4					62.6						.014						
2LS-50	2.7	16.1	11.0	0 168	.5	28.9	8.4	153	2.71	15.1	.4	1.4	2.6	9	.9	1.1	2	45	.08	.093		24.6											.1<.0	
STANDARD DS4	6.6	124.0	29.8	3 159	.3	34.4	11.5	774	3.11	23.4	6.0	25.8	3.8	29	5.5	5.0	5.2	76	.53	.096	17	163.7	.59	138	.082	2	1.73	.032	.17	4.0	-28	5.7	1.1.0	<u> 17 6</u>

Sample type: SOIL SS80 60C. Samples beginning 'RE' are Reruns and 'RRE' are Reject Reruns.

All results are considered the confidential property of the client. Acme assumes the liabilities for actual cost of the analysis only.