

MASS TRANSFER DURING HYDROTHERMAL ALTERATION  
AT THE TAUHARA GEOTHERMAL FIELD

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ABSTRACT

Cores recovered from drillhole TH3 (1094m) and THM1 (430m deep) in the Tauhara Geothermal Field, near Wairakei, show extensive hydrothermal alteration. The observed mineral assemblages reflect the prevailing temperatures, fluid compositions and hydrology. Minerals present include epidote, prehnite, laumontite, wairakite, heulandite, calcite, barite, sphene, leucoxene, adularia, albite, pyrite, pyrrhotite, and rare metal-bearing phases. Clay minerals are common, comprising chlorite, smectites, interlayered illite-smectite and discrete illite; where acidic waters occur, kaolinite (and alunite) are present.

Chemical changes in the host rocks have accompanied the observed alteration: silica shows pronounced enrichment relative to the unaltered rocks but alumina has redistributed variably. Magnesium redistribution matches the presence and abundance of chlorite and smectite, and to some extent correlates with iron concentrations in the cores. Na<sub>2</sub>O and CaO are depleted to differing degrees whereas K<sub>2</sub>O is usually enriched in the altered rocks.

Water and sulphur have been added to most altered rocks but TiO<sub>2</sub>, MnO and P<sub>2</sub>O<sub>5</sub> are present in low abundance in all the cores with only the first showing significant variation. Sr, Rb, Ba and V are among the most mobile trace elements but, by contrast, Y, Th and Zr were relatively immobile during alteration.

The behaviour of mercury suggests that it is transported mainly in the vapour phase, being depleted in reservoir rocks that are hotter than about 225°. However, mercury has been added to rocks comprising the shallow Tauhara aquifer that contains condensed steam (and CO<sub>2</sub>), suggesting that measuring Hg concentration in cores and cuttings from drillholes may be a method of locating zones where potentially corrosive acid sulphate and bicarbonate waters occur.

INTRODUCTION

The Tauhara geothermal field forms part of a larger Wairakei system. Cores recovered from four deep and several shallower monitor wells record evidence of extensive fluid/rock interaction. This is most obvious in their widespread hydrothermal alteration mineralogy (Kakimoto, 1983; Kakimoto and Browne, in prep.) but is also evident from the chemical changes that the host rocks have undergone. This paper describes the changes in rock chemistry that have resulted from hydrothermal alteration in cores recovered from two drillholes (TH3 and THM1).

STRATIGRAPHY AND HYDROTHERMAL ALTERATION

Figures 1 and 2 summarise the stratigraphy, lithology and major features of the hydrothermal alteration seen in these two wells. The sequence encountered consists of pyroclastic silicic rocks and lacustrine sediments comprising mainly the Waiora and Huka Falls Formations. There are three aquifers: the Waiora, consisting of the Waiora Formation itself; the Huka Falls Formation pumice breccias, and that hosted by the Wairakei Lapilli Tuff. The interbedded Huka Falls Formation sediments are much less permeable and serve as partial caps. Primary minerals present in these rocks included mainly quartz and andesine with accessory biotite, hornblende, pyroxene, ilmenite, magnetite and titanomagnetite. Glass was once widespread, and occasional lithic fragments were also present. Hydrothermal alteration has produced a suite of secondary minerals (Figures 1 and 2) that can conveniently be grouped into mineral zones that largely reflect the lithology and hydrology of their host rocks (Kakimoto, 1983; Kakimoto and Browne, in prep.).

COMPOSITION OF CORES FROM DRILLHOLES TH3 AND THM1

Methods

Whole-rock splits of 20 cores from wells TH3 and THM1, chosen to represent the different alteration zones, were analysed for major and trace elements by X-Ray fluorescence (Table 1). Structurally-bound (H<sub>2</sub>O\*) and adsorbed (H<sub>2</sub>O\*) water contents, particle density and porosity were also determined (Table 2). Mercury contents were measured using a Jerome Gold Film mercury detector following the method described by McMerney et al. (1972). Sb, Sn and Ag analyses were made by flame atomic adsorption. Kakimoto (1983) gives a complete account of the analytical methods used in this paper.

Mass transfer of elements between rocks and fluids is most commonly measured as changes in the amount of an element moved per unit volume of rock, relative to its initial (i.e. unaltered) composition (Steiner, 1977; Bogie and Browne, 1979; Henneberger, 1983). This method takes into account changes in rock density resulting from hydrothermal alteration either by replacement or mineral precipitation into cavities. It assumes that prior to their alteration the rocks from the same formation had the same compositions and densities. Reference samples of tuffs (Tables 1 and 2) are of unaltered cores from drillhole THM1-11m and Wairakei drillholes (Steiner, 1977); their compositions and densities were used to estimate net addition and removal of elements in the tuffaceous samples from wells TH3 and THM1 (Table 2).

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	Unaltered <sup>b</sup>										IH3				M3				THM1			
											Micaceous Clay zone				Calcite zone				Epidote Zone			
	Smeectite zone 116 (depth m)																					
THM1-11	W. Tuff	Ig. 11	176	239	361	420	452	482	574		Calcite Zone	Ca (t-s)	Illite Zone	Micaceous Clay Zone	Smeectite Zone	Epidote Zone	Illite Zone	Silicified Zone				
Depth (m)																						
SiO <sub>2</sub> (wt.%)	69.77	67.7	72.4	59.63	61.06	69.32	66.83	72.08	70.42	75.08	70.83	77.56	72.06	76.40	73.40	14.87	60.99	67.03	73.46	62.93	82.90	
TiO <sub>2</sub>	0.33	0.56	0.27	0.58	0.59	0.36	0.27	0.27	0.32	0.32	0.35	0.21	0.41	0.22	0.28	0.25	0.42	0.25	0.36	0.56	0.18	
Al <sub>2</sub> O <sub>3</sub>	13.50	15.1	14.0	23.83	17.81	15.25	14.78	14.64	13.56	12.78	13.93	12.20	13.76	12.27	13.47	13.20	12.78	0.86	13.80	20.50	7.97	
Fe <sub>2</sub> O <sub>3</sub> <sup>c</sup>	2.85	3.0	2.1	1.98	4.31	6.08	2.96	2.50	1.31	2.43	2.94	1.84	2.78	1.48	2.16	1.80	3.49	2.43	2.15	3.64	1.81	
MnO	0.11	0.11	0.09	0.03	0.04	0.15	0.07	0.13	0.07	0.11	0.10	0.09	0.10	0.08	0.12	0.10	0.11	0.07	0.09	0.07	0.05	
HgO	0.49	0.5	0.4	0.56	0.89	2.00	1.04	0.98	0.83	0.61	1.00	0.21	0.62	0.26	0.44	0.18	0.89	1.13	0.48	0.64	0.26	
CaO	2.20	2.7	1.8	0.43	3.86	4.12	1.80	0.91	1.06	2.25	0.58	0.92	2.56	1.48	2.55	1.37	2.39	1.27	2.04	0.84	0.36	
Na <sub>2</sub> O	3.81	4.2	4.0	0.27	2.97	3.12	2.06	0.78	3.45	0.20	0.20	3.04	3.61	5.05	3.21	3.97	3.56	1.71	3.16	5.62	4.81	
K <sub>2</sub> O	2.49	2.8	3.6	1.18	1.28	1.27	1.10	0.97	1.93	3.52	2.78	3.09	2.52	1.95	3.71	3.56	2.68	1.71	3.16	5.62	4.81	
P <sub>2</sub> O <sub>5</sub>	0.08	0.08	0.03	0.04	0.09	0.11	0.10	0.04	0.08	0.10	0.10	0.05	0.09	0.05	0.07	0.05	0.07	0.07	0.08	0.54	0.05	
S	0.01	trace	trace	0.02	1.24	0.51	0.10	0.36	0.06	0.19	0.14	0.17	0.19	0.10	0.19	0.03	0.27	0.63	0.05	0.59	0.26	
H <sub>2</sub> O <sup>+d</sup>	3.86	2.43	0.88	12.28	6.30	3.82	5.10	7.74	3.40	4.10	3.96	3.82	1.36	2.18	1.32	0.86	11.64	10.08	1.88	5.18	0.42	
H <sub>2</sub> O <sup>-e</sup>	0.82	0.71	0.06	3.68	2.79	0.62	1.95	3.42	1.15	1.08	0.64	0.04	0.06	0.04	0.02	0.06	4.97	5.62	0.30	0.65	0.08	
Total <sup>f</sup>	100.32	99.89	99.63	100.49	100.71	101.26	101.21	99.71	100.35	98.77	100.62	100.92	100.80	100.35	100.37	100.17	100.33	100.70	100.15	100.49	101.06	100.35
S = 0	0.004			0.01	0.46	0.19	0.04	0.13	0.02	0.07	0.05	0.06	0.07	0.04	0.07	0.01	0.02	0.10	0.24	0.02	0.22	0.10
Total <sup>f</sup>	100.32			100.48	100.25	101.07	101.17	99.58	100.33	98.70	100.57	99.86	100.73	100.81	100.30	100.76	100.31	100.60	99.91	100.47	100.84	100.25
Rb (mg/kg)	89			48	51	51	43	64	44	213	120	61	113	93	94	131	127	133	77	110	347	198
Sr	172			45	261	259	136	169	141	116	80	193	128	197	110	147	143	382	131	173	25	43
Ba	606			273	460	418	656	630	825	519	519	556	638	512	745	810	710	687	451	712	127	310
V	13			215	145	130	18	38	6	14	18	36	6	28	4	12	8	46	34	19	22	5
Y	30			12	16	20	26	33	31	34	30	18	27	18	24	21	25	19	14	26	66	21
Th	9			8	5	6	10	10	10	7	7	3	10	7	10	6	9	4	4	6	17	2
Zr	186			292	107	129	176	195	212	180	167	143	150	178	179	169	176	176	91	174	330	96
Hf	2			26	17	28	nd	4	nd	nd	1	14	2	4	3	nd	nd	8	7	1	1	1
Cr	22			192	72	115	14	12	17	14	14	84	29	40	30	25	27	40	36	25	9	24
Cu	4			38	18	20	4	16	2	4	3	5	2	4	2	nd	3	10	7	4	9	16
Zn	67			36	60	66	57	50	40	70	56	44	34	47	40	42	38	50	40	52	54	33
Pb	17			14	19	11	12	23	15	11	13	12	11	7	9	9	12	7	7	11	21	7
Sb <sup>g</sup>	<5			<5	nd	nd	nd	nd	<5	nd	nd	nd	nd	<5	<5	<5	<5	<5	nd	nd	nd	nd
Sn <sup>g</sup>	nd			nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd
Ag	nd			nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd
Au <sup>h</sup>	nd			nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd	nd
As <sup>h</sup>																						
Hg (μg/kg) <sup>i</sup>	1774			520	80	82	118	402	35	36	16	11	9	9	26	10	15	550	414	31	369	8
Rock Type	tuff	tuff	ignim- brite	sand- stone	silt- stone	sand- stone	tuff	mudstone	tuff	tuff	tuff	tuff	tuff	tuff	tuff	ignim- brite	tuff	tuff	tuff	tuff	tuff	tuff

TABLE 1: Chemical analyses of selected cores from drillholes TH3 and THM1 compared with unaltered cores from Wairakei (W. Tuff and Ig. 11).

Analyses provided by XRF, unless noted. - = not determined; nd = not detected; sample descriptions and hydrothermal mineralogy listed in Kakimoto, 1983. b. Analyses for unaltered sample from Kakimoto, 1983 (THM1-11), and from Steiner (1977): W. Tuff = Pumiceous and Vitric Tuff, sample No. 11; Ig. 11 = Pumiceous Ignimbrite 11, sample No. 30. e. All iron reported as Fe<sub>2</sub>O<sub>3</sub>, d. Penfield determination. e. Loss upon drying at 110°C. f. Corrected for Sulphur = Oxygen. g. Sb, Sn and Ag analyses by flame AA method, provided by Patricia Hardman, Geology Dept., University of Auckland. h. Au, As (and Sn-detection limit = 1 μg/kg) analyses by emission spectroscopy for TH3-482, TH3-1065 (below detection limits - not shown) and TH3-1094; provided by Terry Seward, Chemistry Division, DSIR, Lower Hutt. i. Hg analyses by gold film detector.

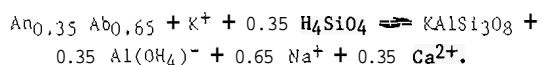
## MAJOR ELEMENTS

## Enrichment and Depletion Trends

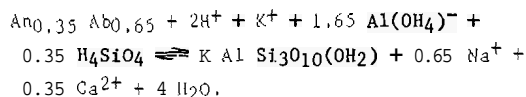
Table 2 and Figures 3 and 4 summarise the mass changes that have occurred in response to the hydrothermal alteration.

Silica shows a net gain, with the most pronounced enrichment being in the silicified zone of THM1 at 390m. Silica addition mainly reflects precipitation of quartz directly from solution and, to a lesser extent, replacement of primary minerals.

Alumina redistribution is variable, with a slight net increase in the calcite zone of TH3, relative immobility in underlying calc-silicate zones and a marked (-80%) increase in the THM1 illite zone (315m).  $Al_2O_3$  is depleted in the clay-free, silicified zone of THM1. These changes result from the transformation of andesine to albite, adularia and calcite which release aluminium, probably in the form of  $Al(OH_4)^-$ . A typical representative reaction is:



The alumina thus released may be fixed locally, or elsewhere (e.g. in the clay-rich zones), as illite according to the reaction:



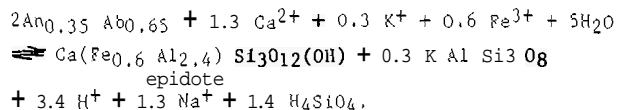
In the calc-silicate zone of TH3, alteration of andesine to calc-silicates (epidote, prehnite, wairakite) results in near isochemical transfer of alumina but whether or not it is added elsewhere depends mainly upon whether or not illite forms.

MgO is enriched in the shallow smectite zone of THM1, being incorporated into the smectite itself, but to a minor extent also into heulandite. Replacement of groundmass and some primary ferro-magnesian minerals by chlorite probably adds MgO to samples from the TH3 calcite zone but its depletion in the underlying calc-silicate zone matches the absence here of interlayered clays and scarcity of chlorite.

The behaviour of iron (expressed as  $Fe_2O_3$  in these analyses) is variable in the two wells but it correlates, to some extent, with the pattern of MgO addition and removal. The enrichment of Fe in THM1-315m (illite zone) correlates with a local abundance of pyrite.

$Na_2O$  and  $CaO$  are depleted to differing degrees in most altered samples whereas  $K_2O$  is usually enriched. The abundance of these elements in rocks and fluids is controlled by mineral equilibria involving hydrolysis ( $H^+$  and  $OH^-$  transfer), hydration and cation-exchange reactions among primary feldspars and hydrothermal minerals.

$K_2O$  is enriched in most samples due to the formation of illite and/or adularia, for example, by the reaction:



Calcium, however, is either depleted from the rocks or is relatively immobile during alteration, being merely reshuffled during the replacement of primary andesine by calcic minerals. Samples containing abundant calcite or calc-silicate minerals (e.g. TH3-482m, 726m, 973m; THM1-66m) show  $CaO$  as being the least depleted with the near isochemical replacement of andesine.

Sodium behaviour also reflects leaching of primary plagioclase in addition to its replacement by albite. In the TH3 calcite zone,  $Na_2O$  concentrations are very low because andesine leaching is extensive; in some THM1 samples,  $Na_2O$  depletion is due to replacement of andesine by smectite and illite.

By contrast, in the TH3 calc-silicate zone, little  $Na_2O$  has been removed from the rocks and, indeed, sample TH3-756m shows a net gain. This correlates with ubiquitous albite present in this zone where it occurs both in the groundmass and as a replacement mineral.

$H_2O^+$  is enriched in the heulandite and clay-rich THM1 smectite zone samples and in the clay-rich calcite zone samples from TH3. Depletion of  $H_2O^+$  in the hotter, TH3 calc-silicate zone may be attributed to dehydration of volcanic glass.

The addition of S during alteration is due to the widespread occurrence of iron sulphides and, more rarely, sulphates in the altered samples.

$TiO_2$ , MnO and  $P_2O_5$  are present in low abundance in all of the rocks. Only  $TiO_2$  shows any significant variation, which mostly follows the pattern for MgO and  $Fe_2O_3$  addition and removal. MnO and  $P_2O_5$  appear to be immobile.

## TRACE ELEMENTS

Sr, Rb, Ba and V are among the more mobile trace elements (Figure 3 and 4); Sr correlates closely with  $CaO$  (Kakimoto, 1983) being highest (382  $\mu g/kg$ ) in core from THM1-63m which contains heulandite. As expected, there is a well-defined linear relationship between Rb and K (Kakimoto, 1983) confirming their chemical similarities. However, the K/Rb ratio of an altered rock commonly reflects its intensity of alteration since adularia has a lower capacity to accommodate Rb than does illite. Thus the lowest K/Rb ratios occur in illite-rich samples such as TH3-482m (300) and THM1-315m (294) whereas samples without illite, but containing adularia, have higher K/Rb ratios, for example THM1-63m (366), 77m (403) and 390m (441).

Barium is invariably enriched in the TH3 samples relative to their unaltered equivalents but THM1 cores show Ba depletion at depth; in part this is due to its positive correlation with potassium for which it can substitute (Roy, 1965, 1969) but barite is also widely present as a trace mineral in many cores.

Y, Th and Zr were relatively immobile during alteration, being mainly fixed in resistate minerals like zircon and apatite, nor do zinc and lead concentration vary significantly; neither galena nor sphalerite were seen. Despite cassiterite, silver minerals and gold being observed, the concentrations of Sn, Ag and Au are below detectable quantities. Antimony was detected in nearly half of the analysed samples but its distribution can not be related to the alteration mineralogy.

Arsenic is present in anomalously high concentrations (0.2 weight %) at the top of the calcite zone of TH3; some may be present within pyrite (up to 200  $mg/kg$ ) but an undetected separate arsenic phase is probably present here too.

## MERCURY

The behaviour of Hg in many geothermal environments suggests that it is transported mainly in the vapour phase as  $Hg^0$  (e.g. Weissberg and Rhode, 1978; Cox, 1983, 1984, 1985; Lynne, 1984; Davey and Van Moort, 1986). In the Tauhara cores, Hg appears to occur predominantly as adsorbed Hg that is volatilised at temperatures above about 200 to 250°C. Mercury enrichment of an altered core should therefore indicate the presence of an Hg-transporting steam phase (or its condensate) and reservoir temperatures below 200-250°.

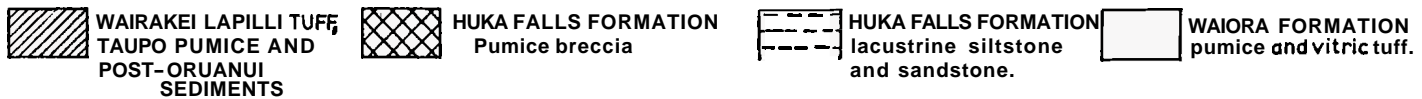
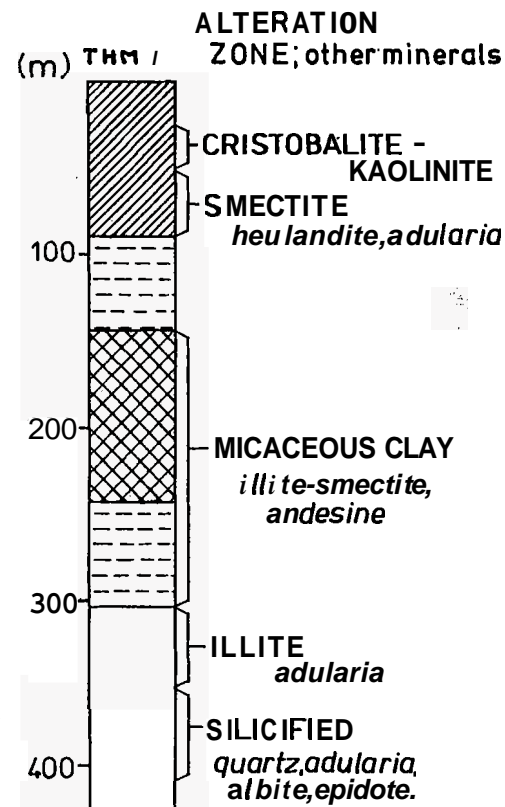
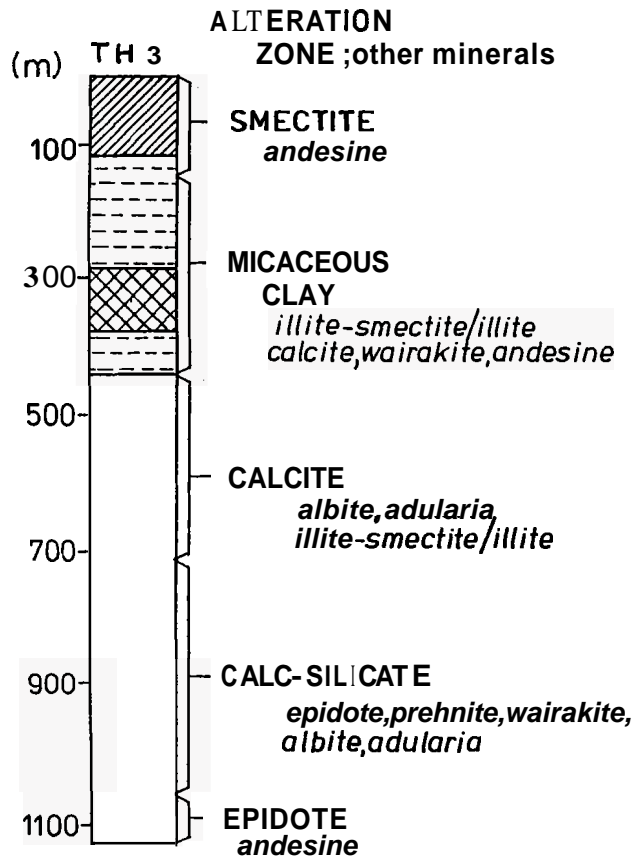


Figure 1: Summary of the distribution of major minerals in alteration zones of drillhole TH3.

Figure 2: Summary of the distribution of major minerals in the alteration zones of THM1.

Sample	Unaltered				TH3										THM1				
	THM1-11 W. Tuff	Tuff Average	19.11	361 (depth m)	152	462	571	615	695	726	756	971	1094	61	77	200	115	390	
SiO <sub>2</sub> (gm/100 cm <sup>3</sup> )	117.1	162	155	174	181.8	181.6	185.8	185.1	186.1	199.1	182.1	191.1	182.8	192.5	155.2	177.5	1117.2	171.5	211.2
TiO <sub>2</sub>	0.7	1.1	1.0	0.65	0.95	0.68	0.84	0.79	0.92	0.51	1.0	0.55	0.70	0.61	1.1	0.66	0.92	1.5	0.16
Al <sub>2</sub> O <sub>3</sub>	28.6	16	12	34	40.4	16.9	36.0	11.5	16.6	11.1	11.8	30.7	11.6	11.9	32.5	16.1	15.2	56.5	20.5
Fe <sub>2</sub> O <sub>3</sub>	6.0	7.2	6.6	5.1	7.8	1.1	6.1	7.2	7.5	1.7	7.0	1.7	5.1	1.6	8.9	6.1	5.5	10.0	1.7
MnO	0.21	0.26	0.24	0.22	0.19	0.18	0.29	0.25	0.24	0.23	0.25	0.20	0.30	0.26	0.28	0.19	0.21	0.19	0.13
MgO	1.0	1.2	1.1	0.96	2.8	2.1	1.6	2.5	1.1	0.54	1.6	0.6	1.1	0.46	2.3	1.0	1.2	1.8	0.67
CaO	1.7	6.5	5.6	1.1	4.8	2.7	5.9	1.4	1.6	2.1	6.5	1.7	6.3	1.5	6.1	1.1	5.2	1.8	0.91
Na <sub>2</sub> O	8.1	10	9.1	9.6	5.5	8.7	nd	0.49	5.2	7.8	9.1	12.6	8.0	10.2	nd	nd	6.7	nd	3.1
K <sub>2</sub> O	5.1	6.7	6.0	8.6	2.9	1.9	9.1	6.8	5.1	7.9	6.1	1.9	9.2	9.2	6.8	1.5	8.0	15.5	12.4
P <sub>2</sub> O <sub>5</sub>	0.17	0.19	0.18	0.07	0.27	0.20	0.21	0.25	0.16	0.13	0.21	0.13	0.17	0.13	0.18	0.19	0.20	0.11	0.13
S	0.02		0.02		0.27	0.15	0.50	0.15	0.15	0.49	0.25	0.17	0.07	0.15	0.69	1.7	0.13	1.6	0.67
H <sub>2</sub> O <sup>a</sup>	8.3	5.8	7.05	2.1	11.5	8.6	12.2	9.8	10.0	1.5	5.5	2.2	1.1	2.2	29.6	26.7	1.8	14.3	1.8
Rb (gm/10 <sup>4</sup> cm <sup>3</sup> )	108				114	111	562	296	161	291	235	207	126	127	338	204	280	957	509
Sr	364				525	155	306	197	508	129	199	212	366	368	972	347	441	69	111
Ba	1283				1607	1587	2177	1280	1463	1610	1296	1640	2017	1826	1719	1194	1814	150	791
V	20				48	15	17	44	95	15	71	9	30	21	117	90	48	61	13
Y	61				69	78	90	71	17	69	46	53	52	61	48	17	66	182	51
Th	19				27	25	18	17	8	26	18	22	15	21	10	11	15	17	5
Zr	391				467	514	175	412	376	386	451	194	421	453	4111	241	143	910	217
Ni	1				nd	nd	nd	2	17	5	10	7	nd	nd	20	18	1	1	3
Cr	46				17	11	17	35	221	75	101	66	62	69	102	95	64	25	62
Cu	8				11	5	11	7	13	5	10	1	nd	8	25	18	10	25	41
Zn	142				151	101	185	138	168	87	119	88	105	9%	127	106	132	149	85
Pb	36				12	38	29	12	12	28	10	20	22	31	18	18	28	58	18
K/Rb <sup>a</sup>	508				465	795	300	126	591	195	191	176	515	508	166	101	522	291	441
ppart (gm/cm <sup>3</sup> ) <sup>b</sup>	2.10	-2.4 <sup>c</sup>		-2.4 <sup>c</sup>	2.60	2.19	2.61	2.15	2.61	2.57	2.51	2.5	2.19	2.57	2.12	2.5	2.54	2.71	2.57
pdry (gm/cm <sup>3</sup> ) <sup>d</sup>	0.72	1.35		1.66	1.56	1.61	1.33	1.63	1.90	1.83	2.01	2	1.98	2.26	1.28	1.0	1.36	1.19	2.46
porosity (%)	64				40	35	19	35	27	29	20	nd	21	12	17	nd	17	49	1

a. Atomic ratio. b. Particle density. c. Estimated particle density. d. Bulk dry density.

TABLE 2: Mass densities of major and trace elements for altered and unaltered tuff samples.

The host rocks of the deep Tauhara aquifer (Figure 5) are significantly depleted in Hg compared with the overlying Huka Falls mudstones that cap it. This distribution is consistent with the high (> 225") temperatures in the deep aquifer which caused Hg to volatilise.

Mercury contents in the TH3 Huka Falls sediments correlate inversely with temperature (Figure 5). These samples are particularly enriched in Hg as it are the mudstones in cores from THM1. This may be due to the fine-grained nature of the sediment as well as the lower temperatures prevailing here (Figure 5). The high Hg levels, nonetheless, indicate the presence of condensed steam (and CO<sub>2</sub>) in the upper Huka Falls mudstones. This conclusion is consistent with the presence of abundant vapour-rich inclusions in sulphate minerals from this zone, and indications drawn from the measured downwell temperature profiles (Kakimoto, 1983). The highest Hg contents occur in the Post Oruanui lake sediments which appear to host a vapour zone (Allis, 1983) and contain an acid alteration assemblage.

By contrast, the shallow cores from TH3 are unaltered, with low Hg contents, not having been affected by steam or steam condensate.

The distribution of Hg in samples from TH3 and THM1 thus reflects reservoir conditions, in particular the location of steam condensate zones and lower reservoir temperatures.

#### CONCLUSIONS

Fluid/rock interaction at Tauhara has produced a suite of hydrothermal minerals that reflect the hydrology and lithology of the reservoir. These mineralogical changes are also evident in chemical changes that the rocks have undergone. Constituents such as SiO<sub>2</sub>, Al<sub>2</sub>O<sub>3</sub>, Na<sub>2</sub>O, K<sub>2</sub>O, CaO and H<sub>2</sub>O<sup>+</sup> redistribute as a result of mineralogical changes, but others (MnO and P<sub>2</sub>O<sub>5</sub>) are largely immobile. Similarly, of the trace elements Sr, Rb and Ba are very mobile but Y, Th and Zr are immobile during the hydrothermal alteration processes that took place here.

Mercury shows a strong tendency to migrate depending upon local reservoir temperature and the identity of the fluid phase. It readily volatilises from its host at temperatures between 200 and 250°C, moving with the steam phase together with CO<sub>2</sub> and H<sub>2</sub>S. Where the steam (and H<sub>2</sub>S and CO<sub>2</sub>) condenses the transported mercury is reabsorbed onto adjacent rocks. Routine analysis of cores and cuttings for mercury while a well is being drilled is, therefore, probably a good way to locate corrosive waters of acid condensate or bicarbonate composition (Hedenquist and Henley, 1985).

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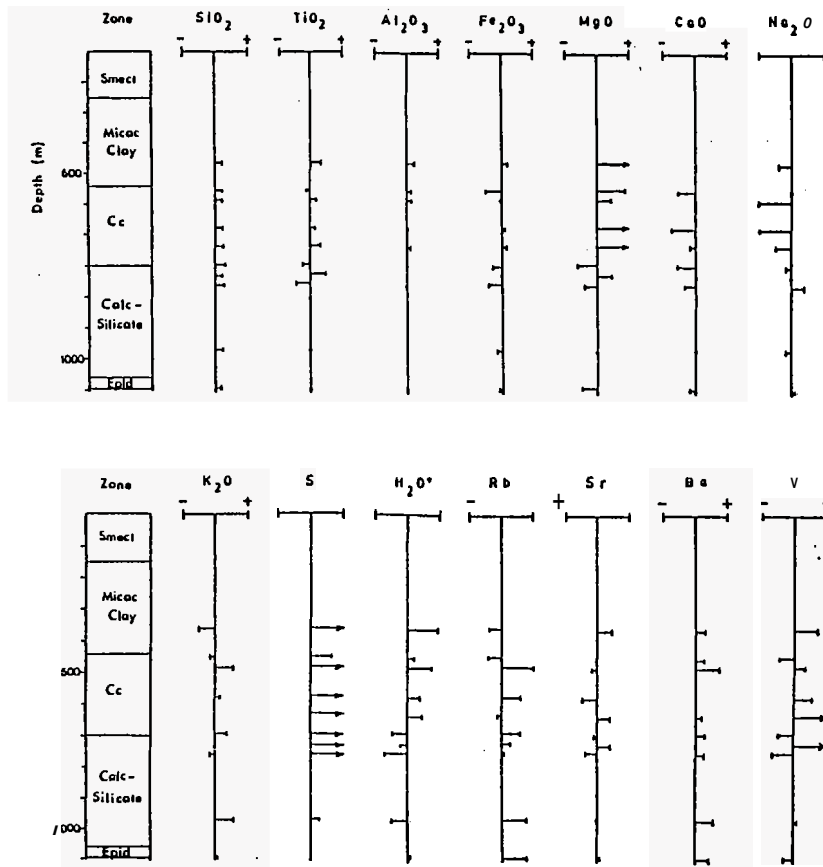


Figure 3: Tuffaceous samples from TH3: percentage change from fresh rock composition for various elements. Scale is  $\pm 100\%$ , except for S which is  $\pm 1000\%$ .

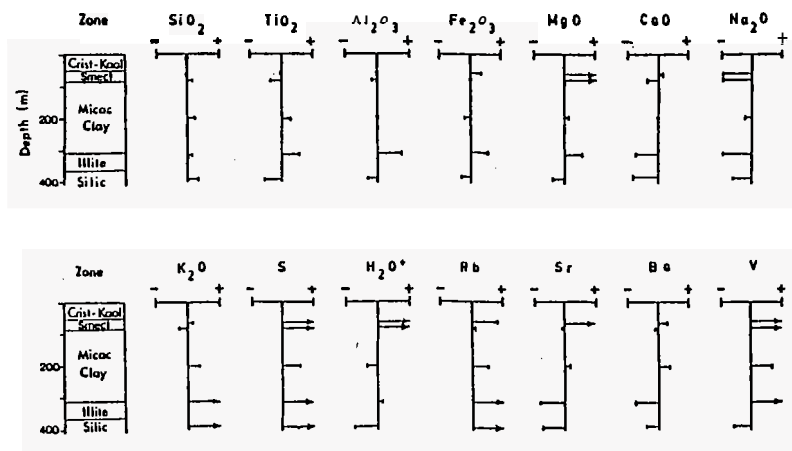


Figure 4: Tuffaceous samples from THM1: percentage change from fresh rock composition for various elements. Scale is  $\pm 100\%$ , except for S which is  $\pm 1000\%$ .

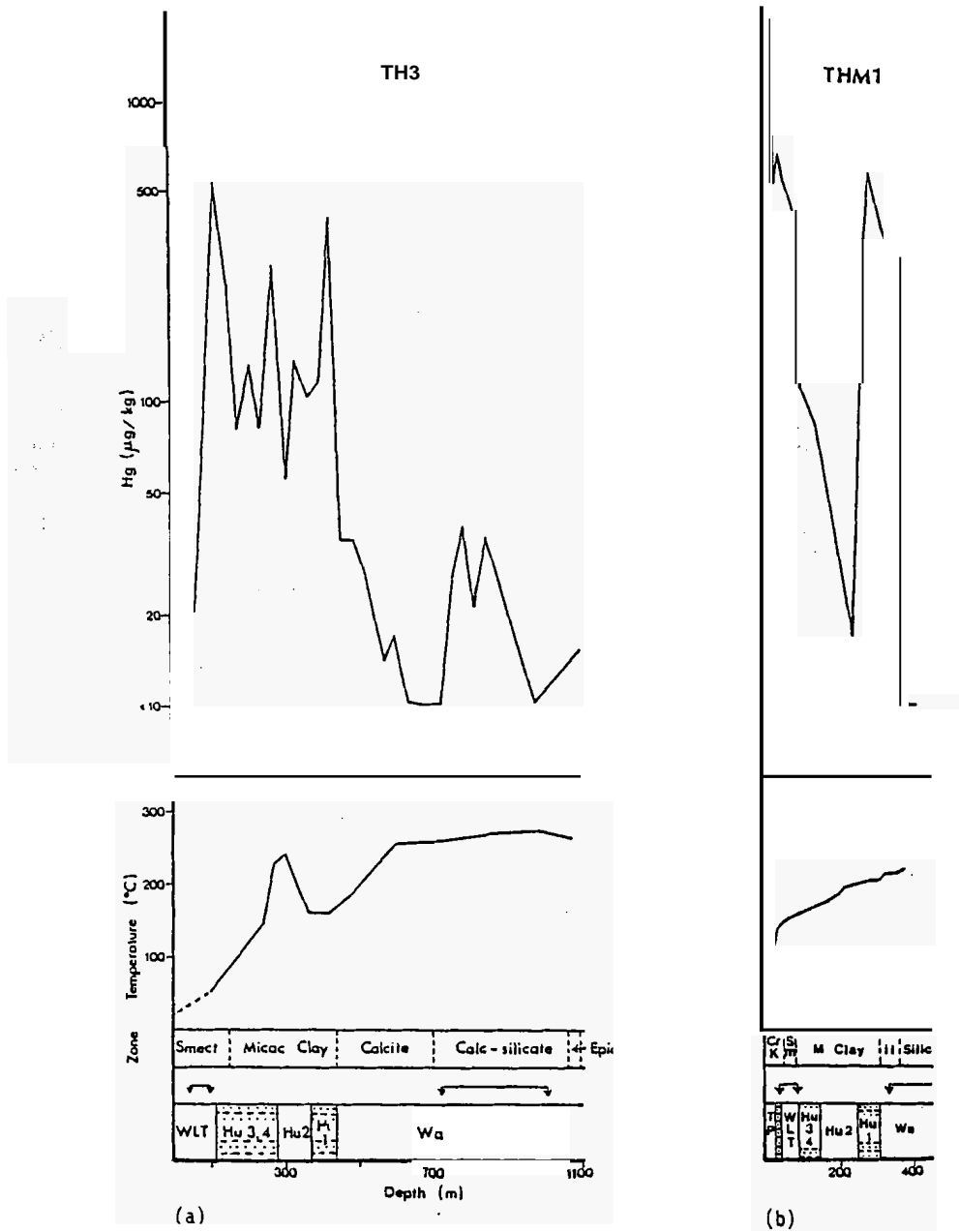


Figure 5: Distribution of Hg with depth in cores from (a) TH3, and (b) THM1; also shown are summary stratigraphy, alteration zones, drillhole temperatures and circulation losses during drilling.