## Chapter 7

## **Reaction Path Modelling of Geological CO<sub>2</sub> Sequestration**

In this chapter we will see how to predict the course of chemical reactions occurring during the geological disposal of  $CO_2$ , building on the thermodynamic and kinetic grounds that were reviewed in previous chapters.

It must be underscored that many early investigations on geological  $CO_2$  storage took into account only trapping of gaseous or supercritical  $CO_2$  as a separated "immiscible" phase under a low-permeability cap rock and  $CO_2$  dissolution into the aqueous phase without considering the reactivity of the aquifer rock, although the latter effect can be very important, especially in the long term as shown below. However, recent studies devoted to geochemical modelling of  $CO_2$  disposal in deep aquifers correctly took into account not only hydrodynamic and solubility trapping but also mineral trapping.

Let us return to suppose we want to know what happens if we inject  $CO_2$  into a deep aquifer. Prior to  $CO_2$  injection, it is reasonable to hypothesize that the aqueous solution stored in the deep aquifer is in (or close to) chemical equilibrium with a certain number of minerals, either the primary rock-forming minerals or, more likely, the authigenic solid phases produced through alteration of the primary minerals at P-T-X conditions different from those at which they formed. Although it makes a lot of difference to refer to the primary minerals rather than to the secondary minerals, here we do not want to reconstruct the evolution of the aquifer rock. Therefore, we simply hypothesize that we know somehow that the deep aqueous solution is in equilibrium with some specified minerals.

## 7.1. The reconstruction of the initial (before $CO_2$ injection) aqueous solution: speciation-saturation calculations

It was shown (e.g., Michard et al., 1981; Chiodini et al., 1991) that the chemical composition of an aqueous solution can be computed, by means of suitable speciationsaturation calculations, if we specify the temperature, the pressure, a certain number of minerals with which the aqueous solution is assumed to be in equilibrium (one for each compatible chemical component, e.g. Na<sup>+</sup>, K<sup>+</sup>, Ca<sup>2+</sup>, Mg<sup>2+</sup>, Al<sup>3+</sup>, SiO<sub>2</sub>),  $f_{CO_2}$ , which fixes HCO<sub>3</sub><sup>-</sup> activity, and the total concentration of each mobile component (e.g., Cl<sup>-</sup>). If multivalent elements, such as Fe, Mn and S, are also present in the considered system, further complications arise. A general, complete account on this subject is given by Van Zeggeren and Storey (1970) and Smith and Missen (1982). But let us proceed step by step.

## 7.1.1. The aqueous solution/calcite example

One of the simplest examples, which is treated in many textbooks, is that of computing the composition of the aqueous solution in equilibrium with calcite at given conditions, e.g. temperature =  $25^{\circ}$ C, pressure = 1 bar and  $P_{CO_2} = 10^{-3.5}$  bar (average atmospheric value). In this case we have to take into account the thermodynamic constants of the dissociation reactions of apparent carbonic acid, bicarbonate ion and water (6-186, 6-187 and 6-188, respectively), i.e.:

$$K_{\rm H_2CO_3^*} = \frac{a_{\rm HCO_3^*} \cdot a_{\rm H^+}}{a_{\rm H_2CO_3^*}} = 10^{-6.35},$$
(7-1)

$$K_{\rm HCO_3^-} = \frac{a_{\rm CO_3^{--}} \cdot a_{\rm H^+}}{a_{\rm HCO_3^-}} = 10^{-10.33},$$
(7-2)

$$K_{\rm H_{2}O} = \frac{a_{\rm OH^-} \cdot a_{\rm H^+}}{a_{\rm H,O}} = 10^{-14.00},$$
(7-3)

respectively, the thermodynamic constant of the dissolution reaction of gaseous  $CO_2$  (6-200):

$$K_{\rm CO_{2(g)}} = \frac{a_{\rm H_2CO_3^*}}{f_{\rm CO_2} \cdot a_{\rm H_2O}} = 10^{-1.47},\tag{7-4}$$

the solubility product of calcite, which is conveniently expressed as follows, referring to the sum of reaction (6-189) and the reverse of reaction (6-187):

$$K_{\text{calcite}} = \frac{K_{\text{sp}}}{K_{\text{HCO}_{3}^{-}}} = \frac{a_{\text{HCO}_{3}^{-}}a_{\text{Ca}^{2+}}}{a_{\text{H}^{+}} \cdot a_{\text{calcite}}} = 10^{+1.85},$$
(7-5)

and the electroneutrality condition

$$2 \cdot m_{\mathrm{Ca}^{2+}} + m_{\mathrm{H}^{+}} = m_{\mathrm{HCO}_{3}^{-}} + 2 \cdot m_{\mathrm{CO}_{3}^{2-}} + m_{\mathrm{OH}^{-}}.$$
(7-6)

Note that we have implicitly made some simplifications, in that we have neglected the formation of complex species, such as CaHCO<sub>3</sub><sup>+</sup> and CaCO<sub>3</sub><sup>o</sup>. If we want to keep calculations to a simple level, further simplifications are needed, namely  $a_{\rm H_2O} = 1$ ,  $a_{\rm calcite} = 1$ ,  $f_{\rm CO_2} = P_{\rm CO_2}$  and  $a_i = m_i$  (i.e.  $\gamma_i = 1$ ) for all solutes. Through suitable substitutions, the

following equation is obtained:

$$\frac{2 \cdot 10^{+1.85} \cdot a_{\rm H^+}^2}{10^{-6.35} \cdot 10^{-1.47} \cdot 10^{-3.5}} + a_{\rm H^+} - \frac{10^{-6.35} \cdot 10^{-1.47} \cdot 10^{-3.5}}{a_{\rm H^+}} - \frac{a_{\rm H^+}}{a_{\rm H^+}} - \frac{2 \cdot 10^{-10.33} \cdot 10^{-6.35} \cdot 10^{-1.47} \cdot 10^{-3.5}}{a_{\rm H^+}^2} - \frac{10^{-14.00}}{a_{\rm H^+}} = 0.$$
(7-7)

This is a fourth-order polynomial equation with respect to  $a_{H^+}$  and its solution without the aid of a computer is a formidable task. By using an electronic spreadsheet we can vary  $a_{H^-}$  until the left-hand side of equation (7-7) changes sign and tends to 0. In this way a pH of 8.26 is obtained. Based on this pH value, the concentrations of other solutes are then easily computed by means of equations (7-1) to (7-5).

Alternatively, we can assume that  $m_{Ca^{2+}} \gg m_{H^+}$  and  $m_{HCO_3^-} \gg 2 \cdot m_{CO_3^{2-}} + m_{OH^-}$ , conditions that have to be verified afterwards. Under these assumptions, equation (7-7) simplifies to

$$\frac{2 \cdot 10^{+1.85} \cdot a_{\rm H^+}^2}{10^{-6.35} \cdot 10^{-1.47} \cdot 10^{-3.5}} - \frac{10^{-6.35} \cdot 10^{-1.47} \cdot 10^{-3.5}}{a_{\rm H^+}} = 0.$$
(7-8)

Equation (7-8) can be easily solved obtaining again pH = 8.26, the same pH value we have found by solving equation (7-7).

Although this coincidence is encouraging, we cannot think to export this approach to more complex problems. In addition, some of the assumptions introduced above may fail: for instance, at high temperature and high ionic strength  $\gamma_i \neq 1$  and  $a_{H_2O} \neq 1$ ; at high pressure  $f_{CO_2} \neq P_{CO_2}$ ; if the considered mineral is not a pure phase its activity is different from unity.

The use of a speciation-saturation code, such as EQ3NR (Wolery, 1992) is mandatory if we want sound results, even in the simple aqueous solution/calcite example. The input file given in Table 7.1 can be used to compute the concentrations and activities of all solute species in an aqueous solution in equilibrium with calcite at temperature =  $25^{\circ}$ C, pressure = 1 bar and  $f_{CO_2} = 10^{-3.5}$  bar. Without entering too much into the details of this code, we are basically saying that: (i) the activity of Ca<sup>2+</sup> ion is fixed by equilibrium with calcite, (ii) the activity of HCO<sub>3</sub><sup>-</sup> ion is governed by equilibrium with a hypothetical, infinite gas reservoir that fixes the  $f_{CO_2}$  of the considered system at  $10^{-3.5}$  bar and (iii) pH is initially set at 8.26 (log  $a_{H^+} = -8.26$ ), but the code has to recalculate it by using the electroneutrality conditions. These are exactly the same conditions used above to compute the approximated pH value, which is now inserted as an input value in EQ3NR (note that it is not necessary to compute an approximated pH value to run this code; EQ3NR is able to run with any pH values or almost so). Note also that, although this exercise has no redox aspect, we set the log oxygen fugacity to -0.678 (the atmospheric value) to avoid possible computational problems.

EQ3NR input file to compute the speciation of the aqueous solution in equilibrium with calcite at 25°C, 1 bar under a  $f_{CO_2}$  of  $10^{-3.5}$  bar (required input formats are not fully respected).

```
Water in equilibrium with calcite at 25 °C and atmospheric P_{\rm CO_2}
Use SUP database
```

endit.

tempc = 2.50	0000E+01								
rho = 1.00	0000E+00	tdspkg	g = 0.0000	00E+00	td	spl = 0.00	000E+00		
fep = -0	.678	uredo	x =						
tolbt = 0.00	0000E+00	told	1 = 0.0000	0E+00	to	lsat = 0.00	000E+00		
itermx= 0									
* 1	2	3	4	5	6	7	8	9	10
iopt1-10 = 0	0	0	0	0	0	0	0	0	0
iopg1-10 = 0	0	0	0	0	0	0	0	0	0
iopr1 - 10 = 0	0	0	1	0	0	0	-1	0	0
iopr11-20 = 0	0	0	0	0	0	0	0	0	0
iodb1 - 10 = 0	0	0	0	0	0	0	0	0	0
uebal = H	-I+								
nxmod =	0								
data file master st	becies = $H^+$								
switch with sr	ecies =								
iflag = 16	csp = -8.26								
data file master sr	ecies = Ca <sup>+</sup>	+							
switch with sr	ecies =								
iflag = 19	csp = 0.								
mineral = Calcite	-op of								
data file master sr	ecies = HCt	0.							
switch with sr	ecies =	- 3							
iflag = 21	csp = -3.50	00							
$mineral = CO_{1}(q)$	-) -)								
endit	/								

The most important parts of the EQ3NR output file are summarized in Table 7.2, which indicates that the pH sought is 8.2748, which is only 0.01 pH units higher than the approximated value computed by means of equations (7-7) and (7-8). In spite of this small difference in pH, Table 7.2 shows that (i) activity coefficients computed by EQ3NR using the extended (B-dot) Debye–Hückel equation (see Chapter 4) deviate somewhat from the value of 1, which was assumed above; these deviations are greater for the divalent ions (e.g.,  $\gamma_{Ca^{2+}} = 0.8445$  and  $\gamma_{CO_3^{2-}} = 0.8451$ ) than for the monovalent ions (e.g.,  $\gamma_{HCO_3^{-}} = 0.9587$  and  $\gamma_{H^+} = 0.9598$ ); (ii) the aqueous complexes CaCO<sub>3</sub>° and CaHCO<sub>3</sub><sup>+</sup> have concentrations of the same order of magnitude of CO<sub>3</sub><sup>2-</sup> ion and, therefore, cannot be neglected.

## 7.1.2. The aqueous solution/multimineral paragenesis general case

A more general case is that of an aqueous solution assumed to be in equilibrium with a paragenesis comprising several solid phases. Again, in this case, the only way to

Part of the EQ3NR output file showing the speciation of the aqueous solution saturated with calcite at 25°C, 1 bar under a  $f_{CO_2}$  of 10<sup>-3.5</sup> bar

Species	Molality	Log molality	Log gamma	Log activity
HCO <sub>3</sub>	9.5399E-04	-3.0205	-0.0183	-3.0388
Ca++	4.8536E04	-3.3139	-0.0734	-3.3874
$O_2(aq)$	2.6518E-04	-3.5765	0.0002	-3.5763
$CO_2(aq)$	1.0739E-05	-4.9691	0.0002	-4.9689
$CO_3^{}$	9.5589E-06	-5.0196	-0.0731	-5.0927
CaCO <sub>3</sub> (aq)	7.0307E06	-5.1530	0.0000	-5.1530
CaHCO <sup>+</sup> <sub>3</sub>	4.3597E-06	-5.3605	-0.0189	-5.3794
OH-	1.9873E-06	-5.7017	-0.0185	-5.7203
<u>H</u> +	5.5335E-09	-8.2570	-0.0178	-8.2748

compute the speciation/saturation state of the aqueous solution is by running a suitable computer code, such as EQ3NR.

If the problem has a redox aspect, we have to assume redox equilibrium, at least if we are using version 7 of the software package EQ3/6, despite most natural systems are far from this ideal condition (e.g., Lindberg and Runnells, 1984), and we have to specify the redox potential of the system, in terms of  $f_{0_2}$ , or Eh, or pe, or through a suitable  $f_{0_2}$ -controlling reaction.

Let us assume that we want to compute the speciation/saturation state of the aqueous solution in equilibrium with analcime, calcite, kaolinite, daphnite, clinochlore, chalcedony and magnetite at 60°C, 1 bar under a  $f_{CO_2}$ , of  $10^{-1}$  bar. Note that this paragenesis include a zeolite, analcime and some phyllosilicates, namely kaolinite, daphnite and clinochlore. The latter two minerals belong to the chlorite group and are assumed to have 14 Å basal spacing. These solid phases are typical of systems of relatively low temperature, as indicated by the relationship between hydrothermal alteration minerals and temperature in active geothermal systems and in hydrothermal ore deposits, representing their fossil analogues (e.g., Henley and Ellis, 1983).

In setting up the EQ3 input file (see Table 7.3), we state that Na<sup>+</sup> activity is fixed by analcime, Ca<sup>2+</sup> activity by calcite, aqueous SiO<sub>2</sub> activity by chalcedony, Al<sup>3+</sup> activity by kaolinite, Mg<sup>2+</sup> activity by clinochlore  $[Mg_5Al_2Si_3O_{10}(OH)_8]$  and Fe<sup>2+</sup> activity by daphnite  $[Fe_5Al_2Si_3O_{10}(OH)_8]$ . Since also magnetite  $[FeO \cdot Fe_2O_3]$  is part of our mineral paragenesis, the  $f_{O_2}$  of the system is uniquely fixed by daphnite–magnetite coexistence, under the redox equilibrium hypothesis. Note that the role of the daphnite–magnetite pair is explicitly declared in the input file (Table 7.3) by writing that daphnite controls Fe<sup>2+</sup> activity and that magnetite constrains  $O_{2(g)}$  activity (incidentally, we have also to set the switch iopt1 to -3, to follow EQ3 instructions).

Similar to the aqueous solution/calcite example: (i) the activity of  $HCO_3^-$ , ion is fixed setting  $f_{CO_2}$ , although a value of  $10^{-1}$  bar is chosen in this case, and (ii) pH is initially set at 7.00, but the code is instructed to recompute it based on the electroneutrality conditions. In addition, total chloride concentration is assumed to be 0.03 mol kg<sup>-1</sup>.

EQ3NR input file to compute the speciation/saturation state of the aqueous solution in equilibrium with analcime, calcite, kaolinite, daphnite, clinochlore, chalcedony and magnetite at 60°C, 1 bar under a  $f_{CO_2}$  of  $10^{-1}$  bar (required input formats are not fully respected).

omissis										
endit.										
tempc=	60.0000E+00									
rho=	1.00000E+00		tdspkg	= 0.0000	00E+00	td	lspl = 0.0	0000E+0	0	
fep=	0.00000E+00		uredox	=						
tolbt=	0.00000E+00		toldl	= 0.0000	00E+00	to	lsat = 0.0	0000E+0	00	
itermx =	0									
*	1	2	3	4	5	6	7	8	9	10
iopt1-10 =	-3	0	0	0	0	0	0	0	0	0
iopg1-10=	0	0	0	0	0	0	0	0	0	0
iopr1-10=	0	0	0	0	0	0	0	0	0	0
iopr11-20 =	0	0	0	0	0	0	0	0	0	0
iodb1-10=	0	0	0	0	0	0	0	0	0	0
uebal = H	<b>I</b> +									
nxmod=	0									
data file master	species = Na	+								
switch with	species=									
iflag = 19	csp = 0.									
mineral = Ana	lcime									
data file master	$species = A1^+$	++								
switch with	species=									
iflag = 19	csp = 0									
mineral = Kao	linite									
data file master	species = Cl <sup>-</sup>									
switch with	species =									
iflag=0	csp = 3.00	000E-02								
data file master	species = SiC	) <sub>2</sub> (ag)								
switch with	species=	21 1/								
iflag= 19	csp = 0.									
mineral = Cha	lcedonv									
data file master	species = $H^+$									
switch with	species=									
iflag= 16	csp = -7.	0000								
data file master	species = $Fe^+$	+								
switch with	species=									
iflag = 19	csp = 0.									
mineral = Dap	hnite-14A									
data file master	$species = O_{2}$	g)								
switch with	species=	.6)								
iflag = 19	csn = 0									
mineral = Mag	metite									
data file master	$species = Ca^{\dagger}$	++								
switch with	species=									
iflag = 10	csp = 0									
mineral = Cal	vite									
mineral – Cal										

```
TABLE 7.3 (Continued)
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```
data file master species = Mg^{++}
switch with species =
jflag = 19 csp = 0.
mineral = Clinochlore-14A
data file master species = HCO_3^-
switch with species =
jflag = 21 csp = -1.0000
gas = CO_2(g)
endit.
```

Results of the EQ3 run are listed in Table 7.4. The computed equilibrium pH is 7.68. At this pH value,  $Al(OH)_4^-$ ,  $HCO_3^-$  and undissociated  $SiO_{2(aq)}$  are by far the prevailing species of dissolved Al, carbonate and silica. Again, the  $MeCO_3^\circ$  and  $MeHCO_3^+$  complex species cannot be neglected, not only for Ca (as in the previous aqueous solution/calcite example), but also for divalent Fe and Mg. In contrast, the free ions Cl<sup>-</sup>, Na<sup>+</sup> and K<sup>+</sup> are by far the prevailing dissolved species of these chemical components.

The code also indicates that the aqueous solution is saturated (i.e.  $-0.3 < \log[Q/K] < +0.3$ , but these limits can be reset by the user) with respect to several minerals, comprising aragonite, Mg- and Na-beidellite, böhmite,  $\alpha$ -cristobalite, 7 Å-cronstedtite, diaspore, huntite, paragonite, quartz, 14 Å-ripidolite, siderite and trydimite. The solid phases with which the aqueous solution is oversaturated and undersaturated are also listed.

Finally, the log  $f_{O_2}$ , resulted to be -61. Probably the log  $(f_{H_2}/f_{H_2O})$  ratio, which is equal to -5.02 in this example, would be a better choice to describe redox conditions, for the reasons given by Giggenbach (1987).

If we also want to include S species into the geochemical model, we may add either pyrite  $[FeS_2]$  or anhydrite  $[CaSO_4]$  or both minerals to the input file. If we choose the pyrite constraint, total S concentration results to be 0.013 mmol kg<sup>-1</sup> and the aqueous solution turns out to be strongly undersaturated with anhydrite. At equilibrium with anhydrite, total S concentration is 480 mmol kg<sup>-1</sup> and the aqueous solution is strongly oversaturated with pyrite. At saturation with both pyrite and anhydrite, total S concentration results to be 472 mmol kg<sup>-1</sup>. In all three cases, however, the log  $f_{0_2}$  of the system is constrained by daphnite-magnetite coexistence at -61. Looking at the thermodynamic constant of the HS<sup>-</sup> -SO<sub>4</sub><sup>2-</sup> equilibrium reaction, written in logarithmic form:

$$\log K_{\rm HS^{-}/SO_{4}^{2-}} = \log a_{\rm SO_{4}^{-}} - \log a_{\rm HS^{-}} - pH - 2 \cdot \log f_{\rm O_{2}}, \tag{7-9}$$

it is evident that pH differences are expected between the pyrite case and the anhydrite case. Indeed, the computed pH is 7.68 in the pyrite case and 6.85 in the anhydrite case (and 6.85 again in the pyrite plus anhydrite case).

Part of the EQ3NR output file showing the speciation of the aqueous solution in equilibrium with analcime, calcite, kaolinite, daphnite, clinochlore, chalcedony and magnetite at 60°C, 1 bar under a  $f_{CO_2}$  of  $10^{-1}$  bar

omissis					
	pH	Eh	pe		
modified NBS pH scale	7.6822	-0.2918	-4.4142E+00		

... omissis

## --- Major Aqueous Species Contributing to Mass Balances ----

Aqueous species accounting for 99% or more of Al+++

Species	Factor	Molality	Per Cent
$\overline{\text{Al}(\text{OH})_4^-}$	1.00	2.3351E-07	97.40
NaAl(OH) <sub>4</sub> (aq)	1.00	3.5248E-09	1.47
Al(OH) <sub>3</sub> (aq)	1.00	2.6892E-09	1.12
Total		2.3973E-07	100.00

Aqueous species accounting for 99% or more of Ca++

Species	Factor	Molality	Per Cent	
Ca <sup>++</sup>	1.00	2.7410E-05	62.86	_
CaHCO <sub>3</sub> <sup>+</sup>	1.00	8.5013E-06	19.50	
CaCO <sub>3</sub> (aq)	1.00	7.6103E-06	17.45	
Total		4.3605E-05	99.81	

Aqueous species accounting for 99% or more of Cl-

Species	Factor	Molality	Per Cent
Cl <sup>-</sup>	1.00	2.9671E-02	98.90
NaCl(aq)	1.00	3.2/3/E-04	1.09
Total		3.0000E-02	100.00

Aqueous species accounting for 99% or more of Fe++

Species	Factor	Molality	Per Cent
FeHCO <sub>3</sub> <sup>+</sup>	1.00	3.0519E-06	79.01
FeCO <sub>3</sub> (aq)	1.00	5.3516E-07	13.86
Fe <sup>++</sup>	1.00	2.7052E-07	7.00
Total		3.8625E-06	99.87

Aqueous species accounting for 99% or more of $HCO_3^-$				
Species	Factor	Molality	Per Cent	
HCO <sub>3</sub>	1.00	5.3742E-02	92.76	
NaHCO <sub>3</sub> (aq)	1.00	2.1026E-03	3.63	
CO <sub>2</sub> (aq)	1.00	1.6092E-03	2.78	
Total		5.7936E-02	99.17	
Aqueous species accoun	ting for 99% or more of Mg	<b>5</b> ++		
Species	Factor	Molality	Per Cent	
Mg <sup>++</sup>	1.00	1.0965E-04	68.44	
MgHCO <sub>3</sub> <sup>+</sup>	1.00	3.8034E-05	23.74	
MgCO <sub>3</sub> (aq)	1.00	1.1295E-05	7.05	
Total		1.6022E-04	99.22	
Aqueous species accoun	ting for 99% or more of Na	+		
Species	Factor	Molality	Per Cent	
Na <sup>+</sup>	1.00	8.3906E-02	97.14	
NaHCO <sub>3</sub> (aq)	1.00	2.1026E-03	2.43	
Total		8.6379E-02	99.57	
Aqueous species accoun	ting for 99% or more of SiC	D <sub>2</sub> (aq)		
Species	Factor	Molality	Per Cent	
SiO <sub>2</sub> (aq)	1.00	5.8790E-04	95.45	
$NaH_3SiO_4(aq)$	1.00	1.5906E-05	2.58	
$H_3SiO_4^-$	1.00	1.2131E-05	1.97	
Total		6.1593E-04	100.00	
omissis				
Mineral	Log Q/K	Aff, kcal	State	
Albite	0.381	0.581	ssatd	
Albite high	-0.735	-1.120		
Albite low	0.381	0.581	ssatd	
Amesite–14A	-2.342	-3.570		
Analcime	0.000	0.000	satd	
Analcime-dehv	-5.602	-8.541	Julu	
Andalusite	-4.423	-6742		
Anthophyllite	-5 945	-9.063		
Antigorite	6 368	9 708	bteas	
Aragonite	-0 143	-0.218	ssaiu	
Artinite	-4 258	-6 492	satu	
Artinite	-4.258	-6.492		

## TABLE 7.4 (Continued)

TABLE 7.4 (Contin	ued)	
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Mineral	Log Q/K	Aff, kcal	State
Beidellite-Ca	-0.408	-0.621	
Beidellite-H	-1.308	-1.993	
Beidellite-Mg	-0.208	-0.317	satd
Beidellite-Na	-0.162	-0.247	satd
Boehmite	-0.315	-0.480	satd
Brucite	-3.237	-4.934	
Graphite	-1.862	-2.839	
Calcite	0.000	0.000	satd
Chalcedony	0.000	0.000	satd
Chamosite-7A	-2.207	-3.364	
Chrysotile	-0.367	-0.559	
Clinochlore-14A	0.000	0.000	satd
Clinochlore-7A	-3.133	-4.776	
Clinoptilolite-Ca	-1.610	-2.455	
Clinoptilolite-Na	1.534	2.338	ssatd
Clinoptilolite-hv-Ca	-1.782	-2.717	
Clinoptilolite-hy-Na	1.545	2.355	ssatd
Coesite	-0.488	-0.744	
Corundum	-3.208	-4.891	
Cristobalite(alpha)	-0.239	-0.364	satd
Cristobalite(beta)	-0.604	-0.921	5
Cronstedtite_7A	-0.062	-0.095	satd
Daphnite_14A	0.002	0.000	satd
Daphnite_7A	-3.107	-4 737	butta
Dawsonite	-0.366	-0.557	
Diaspore	0.036	0.056	satd
Diopside	-3 398	-5 180	butu
Dolomite	1 981	3.019	btezz
Dolomite_dis	0.662	1.009	ssatd
Dolomite_ord	1 981	3 020	ssatd
Enstatite	-1 931	-2 944	
Enidote	-5.113	-7 794	
Epidote-ord	-5.115	-7.797	
Favalite	-2.702	-4 119	
Fe(OH).	-3777	-5 758	
$Fe(OH)_2$	-5.075	-7.736	
FeO	-3.200	-4 878	
Ferrosilite	-1.155	-1 761	
Forsterite	-5 181	-7 898	
Gaulueeite	-5.018	-9.022	
Cibbaita	-0.711	-1.084	
Goothite	-0.711	-0.563	
Greenslite	-0.370	-1 225	
Unlite	-0.810	- 1.235	
Hadanhargita	-4.454	-0.760	
Hoomotito	-4.034	- 7.400	peat-d
Haemaine	0.300	0.379	ssatu
Hercynite	-3.700	-5./51	a a t J
Huntite	0.269	0.411	sata
Hydromagnesite	-6.282	-9.577	
Ice	-0.287	-0.437	satd

Mineral	Log Q/K	Aff, kcal	State
Jadeite	-1.493	-2.276	
Kaolinite	0.000	0.000	satd
Kyanite	-4.232	-6.451	
Lansfordite	-2.868	-4.372	
Laumontite	-2.181	-3.325	
Lawsonite	-3.304	-5.037	
Magnesite	0.541	0.824	ssatd
Magnetite	0.000	0.000	satd
Mesolite	1.666	2.540	ssatd
Minnesotaite	0.489	0.745	ssatd
Monohydrocalcite	-0.977	-1.489	
Montmor-Ca	0.728	1.110	ssatd
Montmor-Mg	0.986	1.503	ssatd
Montmor-Na	1.033	1.575	ssatd
Mordenite	-0.865	-1.318	
Na <sub>2</sub> CO <sub>3</sub>	-6.512	-9.927	
Nahcolite	-2.745	-4.185	
Natrolite	-1.264	-1.927	
Nepheline	-2.821	-4.300	
Nesquehonite	-2.843	-4.335	
Nontronite-Ca	2.242	3.418	ssatd
Nontronite-H	1.343	2.048	ssatd
Nontronite-Mg	2.442	3.722	ssatd
Nontronite-Na	2.488	3.792	ssatd
Okenite	-6.026	-9.187	
Paragonite	0.064	0.097	satd
Pirssonite	-6.073	-9.259	
Prehnite	-5.542	-8.449	
Pseudowollastonite	-5.302	-8.083	
Pyrophyllite	-0.828	-1.262	
Quartz	0.243	0.370	satd
Ripidolite-14A	0.261	0.397	satd
Ripidolite-7A	-2.864	-4.366	
Saponite-Ca	2.462	3.753	ssatd
Saponite-H	1.562	2.381	ssatd
Saponite-Mg	2.657	4.051	ssatd
Saponite-Na	2.707	4.127	ssatd
Scolecite	-0.856	-1.305	
Sepiolite	-2.437	-3.715	
SiO <sub>2</sub> (am)	-0.824	-1.256	
Siderite	0.158	0.241	satd
Sillimanite	-4.727	-7.206	
Talc	2.059	3.140	ssatd
Thermonatrite	-6.532	-9.958	
Tremolite	-3.088	-4.707	
Tridymite	0.087	0.132	satd
Wairakite	-5.593	-8.526	
Wollastonite	-5.118	-7.802	
Wustite	-3.578	-5.455	

TABLE 7.4 (Continued)

omissis							
Gas	Fugacity	Log fugacity					
CH <sub>4</sub> (g)	1.0406E-06	-5.9827					
CO(g)	4.4081E-11	-10.3558					
$CO_2(g)$	1.0000E-01	-1.0000					
$H_2(g)$	1.5491E-06	-5.8099					
$H_2O(g)$	1.6325E-01	-0.7872					
O <sub>2</sub> (g)	1.0378E-61	-60.9839					

TABLE 7.4 (Continued)

#### 7.2. Reaction path modelling

#### 7.2.1. Fundamental relationships

As discussed above, prior to  $CO_2$  injection into the deep aquifer, it is reasonable to hypothesize that the deep aqueous solution is in chemical equilibrium with a certain number of minerals. Upon  $CO_2$  injection, this equilibrium is altered to an extent which depends on the imposed  $f_{CO_2}$  (possible temperature changes caused by isenthalpic  $CO_2$  compression below the Joule-Thompson inversion curve – see Section 3.9 – are neglected in this discussion). As a consequence of this induced disequilibrium, the  $CO_2$ -containing aqueous solution will start to react with the primary minerals of the hosting rock. If the hosting rock is made up of silicates, they will be dissolved and carbonates, silicates, and/or silica minerals will the produced.

Compositional changes during these processes do not depend on the initial and final states only, that is the system can follow different paths in its evolution. In multi-phase systems, different chemical reactions can take place simultaneously and the global chemical evolution can be rather complex. The evolution of the system, possibly towards the final state of chemical equilibrium, is conveniently described by means of a variable named "reaction progress" or "extent of reaction", which is usually indicated by the letter  $\xi$  and is expressed in moles (Prigogine, 1955; Denbigh, 1971; Helgeson, 1979 and references therein). If only one reaction takes place in the system under consideration, the relationship between reaction progress, change in the number of moles of a generic species *i* (e.g., solutes, primary phases destroyed and secondary phases produced),  $dn_i$  and stoichiometric coefficients,  $v_i$  (adimensional), is

$$dn_i = v_i d\xi \tag{7-10}$$

which gives, through integration

$$n_i = n_i^{\circ} + v_i \Delta \xi. \tag{7-11}$$

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The derivative of equation (7-10) with respect to time is (see equation 6-2):

$$\frac{\mathrm{d}n_i}{\mathrm{d}t} = v_i \cdot \frac{\mathrm{d}\xi}{\mathrm{d}t} \tag{7-12}$$

Since the ratio  $d\xi/dt$  represents the rate of the reaction,  $R \pmod{s^{-1}}$ , equation (7-12) can be rewritten as follows:

$$\frac{\mathrm{d}n_i}{\mathrm{d}t} = v_i \cdot R,\tag{7-13}$$

which gives the following equation, through integration:

$$n_i = n_i^{\circ} + v_i R \Delta t, \tag{7-14}$$

under the hypothesis of constant *R*. If the considered reaction is the dissolution of a mineral phase, the rate of the reaction,  $R \pmod{s^{-1}}$ , is related to the dissolution rate,  $r \pmod{m^{-2} s^{-1}}$ , and to the reaction surface,  $A_s \pmod{m^2}$ , by the simple relationship:

$$R = rA_{\rm s}.\tag{7-15}$$

Equation (7-14) can be rewritten as

$$n_i = n_i^{\circ} + v_i r A_s \Delta t, \tag{7-16}$$

Equations (7-11) and (7-16) are very similar. The former describes the increase in the number of moles of component i, with respect to the initial state, based on the reaction progress, whereas the latter makes reference to the time scale. In the second case, it is necessary to define both the rate of dissolution of the solid phase and its reaction surface, as indicated by equation (7-16).

Equations (7-11) and (7-16) are the pivotal relations for reaction path modelling, a technique which has received considerable attention in geochemistry. The thermodynamic relationships describing the irreversible water–rock mass transfers were established by Helgeson (1968). PATHI, the first software code for reaction path modelling, was developed by Helgeson and coworkers, who presented several applications in excellent scientific papers (e.g., Helgeson et al., 1969, 1970). Among the other codes developed afterwards, the EQ3/6 Software Package (Wolery, 1979, 1992; Wolery and Daveler, 1992) is one of the most known and used.

# 7.2.2. An example of reaction path modelling: the dissolution of albite in pure water

As an example, let us take into account the reaction path in a system made up of pure high-albite  $[NaAlSi_3O_8]$  reacting with pure water. This is similar to the K-feldspar example



Figure 7.1. Activity plot for the system  $Na_2O-Al_2O_3$ -SiO<sub>2</sub>-H<sub>2</sub>O, showing two possible reaction paths for highalbite dissolution in pure water, at 25°C, 1 bar, one allowing quartz precipitation (solid grey line), the other preventing it (dashed grey line). Both reaction paths were simulated by means of EQ6.

proposed by Helgeson (1979) in one of his ground-breaking scientific papers. Initially albite and water are in disequilibrium. The activity plot for the system  $Na_2O-Al_2O_3-SiO_2-H_2O$ (Fig. 7.1) shows that the stability fields of kaolinite  $[Al_2Si_2O_5(OH)_4]$ , Na-beidellite  $[Na_{0.33}Al_{2.33}Si_{3.67}O_{10}(OH)_2]$  and paragonite  $[NaAl_3Si_3O_{10}(OH)_2]$  are interposed between those of albite and gibbsite  $[Al(OH)_3]$ .

The model assumes that albite dissolves congruently in pure water, at least in the initial stages, whereas albite dissolution becomes incongruent afterwards, upon attainment of saturation with respect to secondary solid phases. Speciation computations for the initial aqueous solution, containing Na, Al and Si in proportions 1:1:3 (i.e. those of albite), and in low concentrations, show that the dominant species in these initial conditions are Na<sup>+</sup>, Al(OH)<sub>4</sub><sup>-</sup>, Al(OH)<sub>3</sub>° and H<sub>4</sub>SiO<sub>4</sub>°. However, as dissolution of albite proceeds, the concentrations of other product species (e.g., H<sub>3</sub>SiO<sub>4</sub><sup>-</sup>, NaH<sub>3</sub>SiO<sub>4</sub>°, NaOH°, NaAl(OH)<sub>4</sub>°) increase too and have to be taken into account in calculations.

The ratios between the aqueous species produced by albite dissolution remain constant until the solution attains saturation with respect to solid product phases, that will precipitate during the process. This is the basic feature of the reaction path modelling. In order to recognize if the aqueous solution attains saturation with respect to a mineral phase, the solubility products of all the relevant minerals of the considered system have to be compared with the corresponding ion activity products after each increment of the reaction progress variable. In the considered example, gibbsite is the first mineral with which the aqueous solution attains saturation, at point A of Fig. 7.1. The equilibrium between gibbsite and the aqueous solution is described by the following reaction:

$$Al(OH)_{3(s)} = Al^{3+} + 3OH^{-}$$
(7-18)

and the corresponding solubility product is

$$K_{\text{gibbsite}} = a_{\text{Al}^{3+}} \cdot a_{\text{OH}^-}^3.$$
(7-19)

During progressive albite dissolution, gibbsite continues to precipitate and the composition of the aqueous solution evolves along the path A–B in Fig. 7.1. During this process the concentration of dissolved silica (and Na<sup>+</sup>) continues to increase until attainment, at point B, of saturation with respect to kaolinite, which becomes stable and precipitates. Coexistence of gibbsite and kaolinite buffers the activity of aqueous silica, as expressed by the reaction

$$2Al(OH)_{3(s)} + 2SiO_{2(aq)} \rightarrow Al_2Si_2O_5(OH)_{4(s)} + H_2O.$$
(7-20)

Therefore, further albite dissolution does not cause any increment in the concentration of dissolved silica, since this is consumed to convert the gibbsite, precipitated previously, in kaolinite. The Na<sup>+</sup> concentration continues to grow and the net result is path B–C in Fig. 7.1. At point C all the gibbsite has been consumed and the concentration of dissolved silica returns to increase with that of Na<sup>+</sup> following the path C–D along which kaolinite precipitates. At point D, quartz saturation is attained and this mineral starts to precipitate. Further albite dissolution moves the composition of the solution along the path D–E, along which the aqueous silica concentration is buffered by quartz saturation.

At point E the solution attains saturation with paragonite, which begins to precipitate. At point E, the aqueous solution is in equilibrium with quartz, paragonite and kaolinite. Since a maximum of four phases can coexist in a four components system (as stated above, our components are Na<sub>2</sub>O, Al<sub>2</sub>O<sub>3</sub>, SiO<sub>2</sub>, H<sub>2</sub>O) under fixed T, P conditions ( $25^{\circ}$ C, 1 bar in our case), the system has attained an invariant point. The composition of the solution cannot leave point E unless one of the three solids quartz, paragonite and kaolinite is dropped from the system. It must be underscored that albite does not have to be considered in the computation of the phases in equilibrium, since equilibrium between albite and the aqueous solution has not been attained yet and albite acts as supplier of solutes only. Therefore, the composition of the aqueous solution will remain at point E during further albite dissolution, while kaolinite reacts with the aqueous solution to form paragonite according to the reaction

$$1.5Al_{2}Si_{2}O_{5}(OH)_{4(s)} + Na^{+} \rightarrow NaAl_{3}Si_{3}O_{10}(OH)_{2} + 7.5H_{2}O + H^{+}.$$
 (7-21)

Note that, again, minerals' coexistence buffers a parameter of the aqueous solution, this time the ratio between the activities of  $Na^+$  and  $H^+$ .

When kaolinite has been totally consumed, further albite dissolution moves the composition of the solution along the path E–F, along which, again, the concentration of aqueous silica is fixed by quartz saturation. Finally, at point F, albite becomes stable, i.e. the condition of stable equilibrium between albite, paragonite, quartz and the aqueous solution is attained.

Note that, if quartz precipitation were prevented (for instance by kinetic constraints), the composition of the solution would evolve along path D–G. In this case, the condition of metastable equilibrium between albite, Na-beidellite and the aqueous solution would be attained.

The two reaction paths presented above and plotted in Fig. 7.1 simulate what happens if a limited amount of albite is dissolved in the aqueous solution and a constraint is then applied to prevent further dissolution and to allow attainment of equilibrium among all the chemical components. A second dissolution step is then allowed, followed by an equilibration step, and so on. After each step the system reaches a condition of metastable equilibrium for a very small progress of the irreversible reaction. The overall result is a series of metastable equilibrium states extending from the initial state of disequilibrium to the final state of stable equilibrium (point F) or metastable equilibrium (point G). The intermediate states are often named states of partial equilibrium, since the components in the aqueous solution are in mutual equilibrium but in disequilibrium with respect to albite.

#### 7.2.3. The dissolution of albite in pure water: the numeric model

Referring to the albite example, albite dissolution accompanied by gibbsite precipitation is described by means of the following reaction, which includes the main aqueous species of interest:

$$v_{Alb} \text{ NaAlSi}_{3}O_{8(s)} + v_{H_{2}O} H_{2}O \rightarrow v_{Gibb} \text{ Al}(OH)_{3(s)} + v_{Na^{+}} \text{ Na}^{+} + v_{Al^{3^{+}}} \text{ Al}^{3^{+}} + v_{Al(OH)_{3}^{0}} \text{ Al}(OH)_{3}^{0} + v_{Al(OH)_{4}^{-}} \text{ Al}(OH)_{4}^{-} + v_{H_{4}SiO_{4}^{0}} H_{4}SiO_{4}^{0} + v_{H_{3}SiO_{4}^{-}} H_{3}SiO_{4}^{-} + v_{H^{+}} H^{+} + v_{OH^{-}} OH^{-}$$
(7-22)

Actually Al<sup>3+</sup> ion is not a major dissolved species in the considered example but is retained as it is customarily used to describe the dissolution/precipitation of Al-bearing solid phases and the association/dissociation of aqueous Al-species.

Let us assume that the system contains 1,000 g of H<sub>2</sub>O and that the reaction coefficient of albite,  $v_{Alb}$ , is equal to 1. As already seen above, each  $v_i$  represents the change in the number of moles of each species for an increment d $\xi$  of the reaction progress, as expressed by equation (7-10). Reactants have negative  $v_i$  values, whereas the  $v_i$  values of products are positive. In reaction (7-22) we could have erroneously considered as reactants some species that are actually products or vice versa. This is not important since the true situation comes out from computations. Apart from the  $v_{Alb}$ , the other 10  $v_i$  are unknowns and depend on  $\xi$ . To compute these 10 unknowns a system of 10 equations is needed. Let us start to write the thermodynamic equilibrium constant for the five independent reactions between the species involved in reaction (7-22):

$$a_{\mathrm{OH}^-} \cdot a_{\mathrm{H}^+} = K_{\mathrm{W}} \tag{7-23}$$

$$a_{Al^{3*}} \cdot a_{OH^-}^3 = K_{\text{sp,gibbsite}}$$
(7-24)

$$\frac{a_{\rm Al^{3+}} \cdot a_{\rm OH^-}^4}{a_{\rm Al(OH)_4^-}} = K_{\rm Al(OH)_4^-}$$
(7-25)

$$\frac{a_{\rm Al^{3+}} \cdot a_{\rm OH^-}^3}{a_{\rm Al(OH)_3^\circ}} = K_{\rm Al(OH)_3^\circ}$$
(7-26)

$$\frac{a_{\mathrm{H}^*} \cdot a_{\mathrm{H}_3 \mathrm{SiO}_4}}{a_{\mathrm{H}_4 \mathrm{SiO}_4}} = K_{\mathrm{H}_4 \mathrm{SiO}_4}.$$
(7-27)

Now we have to take the derivative of each equation with respect to  $\xi$  and rearrange the results. Let us do this exercise for equation (7-23). The derivative of this relation with respect to  $\xi$  is

$$a_{\rm OH^{-}} \cdot \frac{\mathrm{d}a_{\rm H^{+}}}{\mathrm{d}\xi} + a_{\rm H^{+}} \cdot \frac{\mathrm{d}a_{\rm OH^{-}}}{\mathrm{d}\xi} = 0.$$
(7-28)

Based on equation (7-23), we can substitute  $K_W/a_{H^+}$  for  $a_{OH^-}$  and  $K_W/a_{OH^-}$  for  $a_{H^+}$ , obtaining

$$\frac{K_{\rm w}}{a_{\rm H^+}} \cdot \frac{{\rm d}a_{\rm H^+}}{{\rm d}\xi} + \frac{K_{\rm w}}{a_{\rm OH^-}} \cdot \frac{{\rm d}a_{\rm OH^-}}{{\rm d}\xi} = 0,$$
(7-29)

which can be simplified as follows:

$$\frac{1}{a_{\rm H^{*}}} \cdot \frac{{\rm d}a_{\rm H^{*}}}{{\rm d}\xi} + \frac{1}{a_{\rm OH^{-}}} \cdot \frac{{\rm d}a_{\rm OH^{-}}}{{\rm d}\xi} = 0.$$
(7-30)

Considering that  $m_i = a_i / \gamma_i$ , equation (7-10) can be rewritten in the following form:

$$v_i = \frac{\mathrm{d}m_i}{\mathrm{d}\xi} = \frac{1}{\gamma_i} \cdot \frac{\mathrm{d}a_i}{\mathrm{d}\xi}.$$
(7-31)

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Based on this result and through suitable substitution in equation (7-30), we get

$$\frac{v_{\rm H^{+}} \cdot \dot{\gamma}_{\rm H^{+}}}{a_{\rm H^{+}}} + \frac{v_{\rm OH^{-}} \cdot \dot{\gamma}_{\rm OH^{-}}}{a_{\rm OH^{-}}} = 0$$
(7-32)

or

$$\frac{v_{\rm H^+}}{m_{\rm H^+}} + \frac{v_{\rm OH^-}}{m_{\rm OH^-}} = 0 \tag{7-33}$$

Following the same line of reasoning adopted for equation (7-23), similar relationships are obtained from the expressions of the other four thermodynamic equilibrium constants, equations (7-24) to (7-27):

$$\frac{v_{AI^{3+}}}{m_{AI^{3+}}} + \frac{3 \cdot v_{OH^-}}{m_{OH^-}} = 0$$
(7-34)

$$\frac{v_{Al^{3+}}}{m_{Al^{3+}}} + \frac{4 \cdot v_{OH^-}}{m_{OH^-}} - \frac{v_{Al(OH)_4}}{m_{Al(OH)_4}} = 0$$
(7-35)

$$\frac{v_{Al^{3+}}}{m_{Al^{3+}}} + \frac{3 \cdot v_{OH^-}}{m_{OH^-}} - \frac{v_{Al(OH)_3^0}}{m_{Al(OH)_3^0}} = 0$$
(7-36)

$$\frac{v_{\rm H^{+}}}{m_{\rm H^{+}}} + \frac{v_{\rm H_3SiO_4^{-}}}{m_{\rm H_3SiO_4^{-}}} - \frac{v_{\rm H_4SiO_4}}{m_{\rm H_4SiO_4}} = 0.$$
(7-37)

The other five needed equations are simple mass balances describing the mass exchanges of the five elements involved in reaction (7-22), with respect to albite, under closed system conditions:

Na: 
$$v_{\text{Na}^+} = -v_{\text{Alb}}$$
 (7-38)

Al: 
$$v_{Al^{3+}} + v_{Al(OH)_3^{o}} + v_{Al(OH)_4^{-}} = -v_{Alb}$$
 (7-39)

Si: 
$$v_{H_4SiO_4} + v_{H_3SiO_4^-} = -3 \cdot v_{Alb}$$
 (7-40)

O: 
$$3 \cdot v_{Al(OH)_3^0} + 4 \cdot v_{Al(OH)_4^-} + 4 \cdot v_{H_4SiO_4} + 4 \cdot v_{H_3SiO_4^-} + v_{H_2O} + v_{OH^-} = -8 \cdot v_{Alb}$$
 (7-41)

H: 
$$3 \cdot v_{Al(OH)_3^0} + 4 \cdot v_{Al(OH)_4^-} + 4 \cdot v_{H_4SO_4} + 3 \cdot v_{H_3SO_4^-} + 2 \cdot v_{H_2O} + v_{H^+} + v_{OH^-} = 0.$$
 (7-42)

Equations (7-33) to (7-42) make up the system of 10 equations which allows one to compute the 10 unknown  $v_i$ . The system is conveniently solved through matrix-algebra methods. The reaction progress calculations are initiated by assigning extremely small initial concentrations to the 10 aqueous species involved in reaction (7-22). The system is then solved for the 10 stoichiometric coefficients  $v_i$ . These coefficients are used to compute the new concentrations of all the species after a small reaction step  $\Delta \xi$ , through equation (7-11). It must be underscored that  $n_i$  in equation (7-11) stands for both the molality of aqueous species and the moles of solid phases precipitated from 1 kg of water. For each increment in the reaction progress variable, the speciation of the aqueous solution is computed to evaluate if the aqueous solution has attained saturation with respect to a new solid phase. When a new solid phase is precipitated, the system of equations is suitably modified to incorporate the new products and/or reactants. Since stoichiometric reaction coefficients change along the reaction path, small increments of  $\xi$ , e.g.  $10^{-8}$  mol, have to be used. The best results are obtained modifying equation (7-11), that is including the first derivative of the stoichiometric coefficients with respect to the reaction progress variable:

$$\Delta m_i = v_i \cdot \Delta \xi + \frac{\mathrm{d}v_i}{\mathrm{d}\xi} \cdot \left(\frac{\Delta \xi}{2!}\right)^2. \tag{7-43}$$

This is simply a truncated Taylor's series in  $m_i$ . Equations describing the first derivative of the stoichiometric coefficients with respect to the reaction progress variable,  $dv_i/d\xi$ , are given by Helgeson (1979). These relationships are inserted in the system of equations and at each step of  $\Delta\xi$  the system is solved for  $v_i$  and  $dv_i/d\xi$ . As already said, the reaction progress computation proceeds iteratively, using the concentrations  $m_i$  of each step to calculate the new reaction coefficients  $v_i$ . These, in turn, are used to compute the new concentrations, i.e. the  $m_i$  values pertinent to the subsequent step, and so on. The results obtained through reaction path modelling include the concentrations of all the aqueous species and the number of moles of all the minerals either produced or consumed, for each step of advancement of the reaction progress variable.

The EQ6 input file for albite dissolution is reported in Table 7.5. Results are summarized in Fig. 7.2. Letters A to F in Fig. 7.2 correspond to those in Fig. 7.1. It is clear that both the concentrations of aqueous species and the moles of product minerals vary of several orders of magnitude during the reaction path. The condition of final equilibrium is attained when the total Gibbs free energy of the system  $G = \Sigma \mu_i n_i$  reaches the minimum value. The test concerning this condition is carried out at each step of  $\xi$  and, in the albite example, the minimization of G is achieved for  $\xi = 0.01323$  mol.

## 7.2.4. The dissolution of albite in pure water: simulations in time frame

We have already underscored above the importance of equations (7-11) and (7-16) and their similarity. Equation (7-11) was used to simulate the dissolution of albite through a merely stoichiometric approach (i.e. in reaction progress frame) in the previous section.

EQ6 input file for modeling the dissolution of albite in pure water at  $25^{\circ}$ C, 1 bar in stoichiometric mode (reaction progress frame). Quartz is allowed to form (required input formats are not fully respected).

#### USE the database SUP

The option switch IOPT1 is set to 0 to direct the code to compute the simulation in a reaction progress frame. The option switch IOPT4 is set to 0 to ignore solid solutions.

The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

endit.

nmodl1 = 1			nmodl2:	= 0					
tempc0 = 2.50	jtemp	= 0							
tk1 = 0.00	tk2 = 0.00000E + 00			tk3 = 0.00000E + 00					
zistrt = 0.00	zimax	= 1.00000	E+00						
tstrt = 0.00	000E+00		timemx	= 0.00000	E+00				
kst	pmx=300		cplim=	= 0.00000	E+00				
dzprnt = 0.00	000E+00		dzprig=	= 0.00000	E+00	ksp	pmx=100		
dzplot = 0.10	dzplot = 0.10000E+00			= 0.00000	E+00	ksr	dmx = 1000	)0	
ifile= 60	ifile = 60								
* 1	2	3	4	5	6	7	8	9	10
iopt1-10 = 0	2	0	0	0	0	0	0	0	0
iopt11-20 = 0	0	0	0	0	0	0	0	0	0
iopr1-10 = 0	0	0	1	1	0	-1	1	0	0
iopr11-20 = 0	0	0	0	0	0	0	0	0	0
iodb1-10 = 0	0	0	0	0	0	0	0	0	0
iodb11-20 = 0	0	0	0	0	0	0	0	0	0
nxopt = 1									
option = all									
nxopex = 6									
exception = Gible	bsite								
exception = Kao	linite								
exception = Oua	rtz								
exception = Para	igonite								
exception = Albi	te high								
exception = Beio	lellite-Na								
nffg = 0									
nrct = 1									
*									
reactant= Albi	ite high								
jcode = 0	-		jreac	= 0					
morr = 1.00	000E+01		modr =	= 0.00000	E+00				
nsk = 0			sk	= 0.00000	E+00	fk = 0.00000E + 00			
nrk = 1			nrpk =	= 0					
rk1 = 1.00	000E+00		rk2 =	= 0.00000	E+00		rk3 = 0.00	000E+00	
*									
dlzidp = 0.00	000E+00								
tolbt = 0.00	000E+00		toldl	= 0.00000	E+00		tolx = 0.00	000E+00	
tolsat = 0.00	000E+00		tolsst	= 0.00000	E+00				
screw1 = 0.00	000E+00		screw2:	= 0.00000	E+00	scr	w=0.00	000E+00	
screw4 = 0.00	000E+00		screw5	= 0.00000	E+00	sci	ew6 = 0.00	000E+00	
zklogu = 0.00	)0		zklogl :	= 0.000		zł	dac = 0.00	0	
dlzmx1 = 0.00	000E+00		dlzmx2:	= 0.00000	E+00	nor	dlm = 0		

TABI	E 7.	5 (	Contin	ued)
			~~~~~	V

itermx=	= 0	ntrymx = 0	
npslmx=	= 0	nsslmx = 0	ioscan = 0
* pickup file	written by EO3NR, version 7	7.2b (R139)	<u></u>
* supported l	by EQLIB, version 7.2b (R16)	8)	
Pure water		<i>`</i>	
endit.			
tempci =	2.50000E+01		
nxmod =	0		
iopg1 =	0	iopg2 = 0	iopg3 = 0
iopg4 =	• 0	iopg5 = 0	iopg6 = 0
iopg7 =	: 0	iopg8 = 0	iopg9 = 0
iopg10 =	• 0		
kct =	5	ksq = 6	$kmt \approx 6$
kxt =	6	kdim = 6	kprs $\approx 0$
0	5.550896571687376E+01		
Al	9.9999999999999892E-21		
Н	1.110168703247900E+02		
Na	1.0000000000000E-20		
Si	9.99999999999999974E-21		
Electr	-2.442121511592103E-23		
H <sub>2</sub> O	H <sub>2</sub> O	1.744358983526984E+00	
Al+++	Al+++	2.522092083305848E+01	
H <sup>+</sup>	H <sup>+</sup>	-6.997388938332554E+00	
Na <sup>+</sup>	Na <sup>+</sup>	-2.00000000692422E+01	
$SiO_2(aq)$	$SiO_2(aq)$	-2.000048167543259E+01	
$O_2(g)$	$O_2(g)$	-6.78000000000000E-01	

The same process can be transposed in a time frame by using equation (7-16), which indicates that the rate of dissolution of albite, the rate of precipitation of solid product phases and the reaction surfaces of all minerals, reactant and products, have to be specified to accomplish this task.

The characterization of reaction surface areas of mineral phases is probably the highest difficulty of reaction path modelling, as recognized by many authors in water-rock interaction studies (e.g., Lichtner, 1996, 1998). In particular, the surface of contact between minerals and percolating meteoric waters in soils and rocks may deviate from the surface area under controlled mineral dissolution/precipitation experiments conducted in the laboratory. According to Appelo and Postma (1996), the estimation of mineral surface areas in field situation has hardly passed the stage of educated guessing at present. Marini et al. (2000) have modelled the chemistry of Bisagno Valley (Genoa, Italy) groundwaters, focussing on the main dissolved constituents, using an inverse approach, i.e. introducing the experimental kinetic rate constants of dissolving and precipitating mineral phases and solving for their surface areas. It turns out that computed surface areas of solid phases change substantially during progressive water-rock interaction and differ by several orders of magnitude from those obtained on the basis of modal mineralogy, grainsize and intergranular



Figure 7.2. (a) Log-molalities of dissolved elements and log-activity of  $H^+$  ion and (b) log-moles of secondary phases and of destroyed albite, as a function of the reaction progress variable, during the progressive dissolution of albite in pure water at 25°C, 1 bar, simulated by means of EQ6. Quartz is allowed to precipitate in the simulation. Letters A to F correspond to those in Figure 7.1.

porosity. Thus, even the relative surface areas do not appear to correlate with the modal abundance of minerals in the rocks.

In spite of these difficulties, it is instructive to simulate albite dissolution in kinetic mode, activating quartz precipitation. To keep these runs to a simple level, only the rate of dissolution of albite and its initial reaction surface area are specified in the EQ6 input files of Table 7.6, whereas product minerals are assumed to attain instantaneous equilibrium. This choice is dictated by the difficulty in defining the surface area for solid phases which are initially absent in the system.

In the first input file (see Table 7.6A), the dissolution of albite is described by means of a TST-based rate law, comprising the proton-, water- and hydroxyl-promoted mechanisms, whose rate constants and reaction orders are derived from Table 6.1. Note that the reaction orders in Table 6.1, +0.457 for the acid mechanism, 0.000 for the neutral mechanism and -0.572 for the basic mechanism, are all referred to proton activity, whereas in the EQ6 input file (Table 7.6) the reaction orders have to be referred to H<sup>+</sup>, H<sub>2</sub>O and OH<sup>-</sup> activities, respectively, and consequently they become +0.457, +1.000 and +0.572. In addition, the rate constant for the basic mechanism given in Table 6.1 is multiplied times  $K_W^{-0.572}$ , to refer it to OH<sup>-</sup> activity rather than to H<sup>+</sup> activity.

In the second input file (see Table 7.6B), the dissolution of albite is described by means of the rate law proposed by Oelkers et al. (1994), see equation (6-113). The purpose of this run is to show that this kind of rate laws can also be taken into account by the EQ6 code. Since equation (6-113) holds at pH 9 and 150°C, albite dissolution is simulated in a 1N solution of Na-HCO<sub>3</sub>-CO<sub>3</sub> whose initial speciation, as computed by EQ3NR, is reported below.

Species	Molality	Log molality	Log gamma	Log activity	
Na <sup>+</sup>	9.9941E-01	-0.0003	-0.2462	-0.2464	
HCO <sub>3</sub>	5.9758E-01	-0.2236	-0.2313	-0.4549	
CO3-	1.9873E-01	-0.7017	-0.9534	-1.6552	
OH-	4.3691E-03	-2.3596	-0.2840	-2.6436	
$CO_2(aq)$	1.4647E-03	-2.8343	0.1157	-2.7185	
NaOH(aq)	5.8880E-04	-3.2300	0.0000	-3.2300	
$O_2(aq)$	1.4821E-04	-3.8291	0.1157	-3.7134	
H <sup>+</sup>	1.3149E-09	-8.8811	-0.1189	-9.0000	

Carbonate equilibria and pH constrain CO<sub>2</sub> fugacity at  $10^{-0.6728}$ . This  $f_{CO_2}$  value is imposed on the system during the subsequent EQ6 simulation of albite dissolution, to maintain the pH constant at the desired value of 9.

As shown in Table 7.6B, the dissolution rate of albite is also described in this second case by means of a TST-based rate law, but contrary to the previous case, it includes one mechanism only, involving the activities of hydrogen ion and aluminium ion with exponents 1 and -1/3, respectively, as dictated by equation (6-113). The rate constant, expressed in mol cm<sup>-2</sup> s<sup>-1</sup>, is inserted in the input file as rk0.

#### TABLE 7.6A

EQ6 input file for modeling the dissolution of albite in pure water at  $25^{\circ}$ C, 1 bar in kinetic mode (time frame). Quartz is allowed to form (required input formats are not fully respected).

#### USE the database COM

The option switch IOPT1 is set to 1 to direct the code to compute the simulation in a TIME frame.

The option switch IOPT4 is set to 0 to ignore solid solutions.

The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

endit.

nmodl1=1 nmodl2=0											
tempc0=2.5	60000E+01	jtemp=	=0								
tk1 = 0.0	0000E + 00		tk2=	=0.00000E	+00	tk3=0.00000E+00					
zistrt = 0.00000E + 00 tstrt = 0.00000E + 00			zimax=	=1.00000E	08						
			timemx=	=0.00000E	+00						
kstpmx=300		cplim=	=0.00000E	+00							
dzprnt=0.0	$00000E \pm 00$		dzprlg=	=0.00000E	+00	ksppmx =	ksppmx=100				
dzplot=0.1	0000E + 00		dzpllg=	=0.00000E	+00	ksplmx=	10000				
ifile $= 60$	)										
* 1	2	3	4	5	6	7	8	9	10		
iopt1-10=1	2	0	0	0	0	0	0	0	0		
iopt11-20=0	0	0	0	0	0	0	0	0	0		
iopr1-10=0	0	0	1	1	0	-1	1	0	0		
iopr11-20=0	0	0	0	0	0	0	0	0	0		
iodb1-10=0	0	0	0	0	0	0	0	0	0		
iodb11-20=0	0	0	0	0	0	0	0	0	0		
nxopt=1											
option=all											
nxopex=6											
exception=Gil	bbsite										
exception=Ka	olinite										
exception=Qu	lartz										
exception=Par	ragonite										
exception=Al	bite high										
exception=Be	idellite-Na										
nffg=0											
nrct = 1											
*											
reactant=Al	bite high										
jcode=0			jreac	=0							
morr = 1.0	00000E + 00		modr	=0.00000E	+00						
nsk = 1			sk÷	=1000.00E	2+00	fk=	=0.00000E-	+00			
nrk=2			nrpk	= 1							
imech=3											
rk0=1.3	34900E-14		trk0=	=25.0000E	+00	iact=	=0				
eact=0.0	00000E + 00		hact	=0.00000E	2+00						
ndact=1			csigma	=1.00000E	2+00						
udac=H <sup>+</sup>	F		cdac	=0.45700E	E + 00						

TABI	F7	64	(Continued)
TUDE		י הט.	Commueu

rk0=	=9.12010E-17	trk0=25.0000E+00	iact=0
eact=	=0.00000E+00	hact=0.00000E+00	
ndact=	= 1	csigma=1.00000E+00	
udac=	=H <sub>2</sub> O	cdac = 1.00000E + 00	
rk0=	=1.06660E-13	trk0 = 25.0000E + 00	iact=0
eact=	= 0.00000E + 00	hact = 0.00000E + 00	
ndact=	= 1	csigma=1.00000E+00	
udac=	$=OH^-$	cdac=0.57200E+00	
dlzidp≈	=0.00000E+00		
tolbt≈	=0.00000E+00	told1=0.00000E+00	tolx=0.00000E+00
tolsat=	=0.00000E+00	tolsst = 0.00000E + 00	
screw1=	=0.00000E+00	screw2=0.00000E+00	screw3=0.00000E+00
screw4=	=0.00000E+00	screw5=0.00000E+00	screw6=0.00000E+00
zklogu≈	=0.000	zklog1=0.000	zkfac=0.000
dlzmx1≈	=0.00000E+00	dlzmx2=0.00000E+00	nordlm=0
itermx=	=0	ntrymx=0	
npslmx=	=0	nsslmx=0	ioscan=0
endit. tempci=	=2.50000E+01		
nxmod=	=0	ions2-0	iono2-0
iopg1-	-0	$\log_2 = 0$	lopg5=0
iopg7-	-0	iopg8=0	$\log \theta = 0$
iong10-	-0	lopga=0	10pg9=0
kct=	=5	ksa=6	kmt=6
kxt=		kdim=6	kprs=0
0	5.550896571687376E+01		apis 0
Al	9.99999999999999892E-21		
Н	1.110168703247900E+02		
Na	1.00000000000000E-20		
Si	9.99999999999999974E-21		
electr -	-2.442121511592103E-23		
$H_2O$	H <sub>2</sub> O	1.744358983526984E+	-00
A1+++	Al <sup>+++</sup>	-2.522092083305848E+	-01
$\mathrm{H}^+$	$H^+$	-6.997388938332554E+	-00
$Na^+$	Na <sup>+</sup>	-2.00000000692422E+	-01
SiO <sub>2</sub> (aq)	SiO <sub>2</sub> (aq)	-2.000048167543259E+	-01
$O_2(g)$	O <sub>2</sub> (g)	-6.78000000000000 -	-01

#### TABLE 7.6B

EQ6 input file for modeling the dissolution of albite in a 1 N solution of Na-HCO<sub>3</sub>–CO<sub>3</sub> (pH 9) at 150°C, 4.76 bar in kinetic mode (time frame). Quartz is allowed to form (required input formats are not fully respected).

```
USE the database COM
```

The option switch IOPT1 is set to 1 to direct the code to compute the simulation in a TIME frame.

The option switch IOPT4 is set to 0 to ignore solid solutions.

The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

endit.

ondit.										
nmodl 1 = 1	nmodl2	t = 0								
tempc0 = 1	jtemp	0 = 0								
tk1 = 0.00000E + 00			tk2	= 0.00000	0E+00	tk3=	= 0.00000E	2+00		
zistrt = 0	.00000E + 00		zimax	= 2.30400	0E-03					
tstrt = 0	.00000E + 00		timemx	= 0.00000	0E+00					
kstpmx = 3	00		cplim	= 0.00000	0E+00					
dzprnt= 1	.00000E04		dzprlg	y = 0.00000	0E+00	ksppmx = 100				
dzplot = 0	.10000E + 00		dzpllg	y = 0.00000	0E + 00	ksplmx=	= 10000			
ifile= 6	0									
* 1	2	3	4	5	6	7	8	9	10	
iopt1-10 = 1	2	0	0	0	0	0	0	0	0	
iopt11-20=0	0	0	0	0	0	0	0	0	0	
iopr1-10 = 0	0	0	1	1	0	-1	1	0	0	
iopr11-20=0	0	0	0	0	0	0	0	0	0	
iodb1-10=0	0	0	0	0	0	0	0	0	0	
iodb11-20=0	0	0	0	0	0	0	0	0	0	
nxopt = 1										
option = a	11									
nxopex = 6										
exception = G	libbsite									
exception = K	aolinite									
exception = Q	Quartz									
exception = P	aragonite									
exception = A	lbite high									
exception = B	eidellite-Na									
nffg = 1										
species = C	$O_2(g)$									
moffg = 1	.00000E + 00		xlkffg	s = -0.672	280E+00					
nrct = 1										
*										
reactant = A	lbite high									
jcode= 0			jreac	= 0						
morr = 1	.00000E+00		modi	= 0.00000	0E + 00					
nsk= 1			sk	= 1000.01	E+00	fk=	= 0.00000E	8+00		
nrk = 2			nrpk	x = -1						
imech = 1										
rk0= 2	.90000E-12		trkC	= 150.000	0E + 00	iact=	0			

TABLE 7.6B (Ca	ontinued	
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eact=	= 0.00000E+00	hact = 0.00000E + 00	
ndact=	= 2	csigma = 1.00000E + 00	
udac≈	• H <sup>+</sup>	cdac = 1.00000E + 00	
udac=	= Al <sup>+++</sup>	cdac = -0.33333E + 00	
*		<u>16 17 3</u>	
dlzidp=	= 0.00000E + 00		
tolbt=	= 0.00000E+00	toldl = 0.00000E + 00	tolx = 0.00000E + 00
tolsat=	= 0.00000E + 00	toIsst = 0.00000E + 00	
screw1=	= 0.00000E + 00	screw2 = 0.00000E + 00	screw3 = 0.00000E + 00
screw4=	= 0.00000E+00	screw5 = 0.00000E + 00	screw6 = 0.00000E + 00
zklogu≈	= 0.000	zklogl = 0.000	zkfac = 0.000
dlzmx1=	= 0.00000E+00	dlzmx2 = 0.00000E + 00	nordlm = 0
itermx=	: 0	ntrymx = 0	
npslmx≕	: 0	nsslmx = 0	ioscan = 0
* supported Soluzione a	by EQLIB, version 7.2b (R pH 9 150ØC	168)	
Soluzione a	pH 9, 150ØC		
endit.			
tempci=	= 1.50000E+02		
nxmod=	= () 		
iopg1=	• 0	10pg2 = 0	10pg3 = 0
iopg4=	: ()	10pg5 = 0	10pg6=0
10pg/=	0	10pg8 = 0	10pg9=0
10pg10=	<u> </u>		
10	- 0	1 5	
kct=	= 0 = 6 = 7	ksq = 7	kmt = 7
kct= kxt=	= 0 = 6 = 7 = 5 7005554752122185 + 0	ksq = 7 kdim = 7	kmt = 7 kprs= 0
kct= kxt= O	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140235818E 200</li> </ul>	ksq= 7 kdim= 7	kmt= 7 kprs= 0
kct= kxt= O Al	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.161040002755647E+0</li> </ul>	ksq= 7 kdim= 7	kmt= 7 kprs= 0
kct= kxt= O Al H	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.077766682638062E-01</li> </ul>	ksq= 7 kdim= 7 2	kmt= 7 kprs= 0
kct= kxt= O Al H C No	<ul> <li>0</li> <li>5</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.00000000012328E+0</li> </ul>	ksq= 7 kdim= 7 2	kmt= 7 kprs= 0
kct= kxt= O Al H C Na Si	<ul> <li>0</li> <li>5</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.0000000012290E+0</li> </ul>	ksq= 7 kdim= 7 2 1 0	kmt= 7 kprs= 0
kct= kxt= O Al H C Na Si electr	<ul> <li>0</li> <li>5</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.00000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.600239705566677E-05</li> </ul>	ksq= 7 kdim= 7 2 1 0 0	kmt= 7 kprs= 0
kct= kxt= O Al H C Na Si electr	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.00000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H.O.</li> </ul>	ksq= 7 kdim= 7 11 2 1 0 0 3 1 744358983526984Ed	kmt = 7 kprs = 0
kct = kxt = 0 Al H C Na Si electr H <sub>2</sub> O Al <sup>+++</sup>	<ul> <li>0</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H<sub>2</sub>O</li> <li>A1<sup>+++</sup></li> </ul>	ksq = 7 kdim = 7 1 2 1 1.744358983526984E + - 4.121808857087895E +	kmt = 7 kprs = 0
kct = kxt = 0 Al H C Na Si electr H <sub>2</sub> O Al <sup>+++</sup> H <sup>+</sup>	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H<sub>2</sub>O</li> <li>Al<sup>+++</sup></li> <li>H<sup>+</sup></li> </ul>	ksq= 7 kdim= 7 1 2 1.744358983526984E+ - 4.121808857987895E+ - 8.881099514946308E+	kmt = 7 kprs = 0
kct= kxt= O Al H C Na Si electr $H_2O$ Al <sup>+++</sup> H <sup>+</sup>	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H<sub>2</sub>O</li> <li>Al<sup>+++</sup></li> <li>H<sup>+</sup></li> <li>HCO<sup>-</sup></li> </ul>	ksq= 7 kdim= 7 1 2 1.744358983526984E+ - 4.121808857987895E+ - 8.881099514946308E+ - 2.236025491244946E=	kmt = 7 kprs = 0 -00 -01
kct=kxt=0 Al H C Na Si electr H <sub>2</sub> O Al <sup>+++</sup> H <sup>+</sup> HCO <sub>3</sub> Na <sup>+</sup>	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H<sub>2</sub>O</li> <li>Al<sup>+++</sup></li> <li>H<sup>+</sup></li> <li>HCO<sub>3</sub><sup>-</sup> Na<sup>+</sup></li> </ul>	ksq= 7 kdim= 7 1 2 1.744358983526984E4 - 4.121808857987895E4 - 8.881099514946308E4 - 2.236025491244946E- - 2.557881778253819E-	kmt = 7 kprs = 0 -00 -01 -00 01 04
kct=kxt=0 Al H C Na Si electr H <sub>2</sub> O Al <sup>+++</sup> H <sup>+</sup> HCO <sub>3</sub> Na <sup>+</sup> SiO <sub>4</sub> (aq)	<ul> <li>0</li> <li>6</li> <li>7</li> <li>5.790555475213318E+0</li> <li>1.000000140335818E-20</li> <li>1.116194099975864E+0</li> <li>7.977766682638062E-01</li> <li>1.0000000012290E+0</li> <li>1.000000019167043E-20</li> <li>1.690239705560677E-08</li> <li>H<sub>2</sub>O</li> <li>Al<sup>+++</sup></li> <li>H<sup>+</sup></li> <li>HCO<sub>3</sub><sup>-</sup> Na<sup>+</sup></li> <li>SiO<sub>2</sub>(aq)</li> </ul>	ksq= 7 kdim= 7 11 2 1. 1. 2 1. 1.744358983526984E4 - 4.121808857987895E4 - 4.121808857987895E4 - 8.881099514946308E4 - 2.236025491244946E- - 2.557881778253819E- - 2.116808380506397E4	kmt = 7 kprs = 0 -00 -01 -00 01 04 -01

ξ	Time days	pН	m <sub>Al</sub>	m <sub>C</sub>	m <sub>Na</sub>	$m_{\rm SiO_2}$	$a_{Al^{3}}$	Affinity kcal×mol <sup>-1</sup>	Rate $mol \times cm^{-2} \times s^{-1}$
0.00E+00	0.00E+00	9	1.00E-20	0.798	1.000	1.00E-20	6.05E-42	-135.5676	4.94E-07
1.00E-09	2.34E-11	9	1.00E-09	0.798	1.000	3.00E-09	6.05E-31	-47.6005	1.06E-10
3.16E-09	2.71E-07	9	3.16E-09	0.798	1.000	9.49E-09	1.91E-30	-43.7280	7.25E-11
1.00E-08	1.62E-06	9	1.00E08	0.798	1.000	3.00E-08	6.05E-30	-39.8554	4.94E-11
3.16E-08	8.02E-06	9	3.16E-08	0.798	1.000	9.49E08	1.91E-29	-35.9829	3.36E-11
1.00E-07	3.77E05	9	1.00E-07	0.798	1.000	3.00E-07	6.05E29	-32.1104	2.29E-11
3.16E-07	1.76E-04	9	3.16E-07	0.798	1.000	9.49E-07	1.91E-28	-28.2378	1.56E-11
1.00E-06	8.16E-04	9	1.00E-06	0.798	1.000	3.00E-06	6.05E-28	-24.3653	1.06E-11
3.16E-06	3.79E-03	9	3.16E-06	0.798	1.000	9.49E06	1.91E-27	-20.4928	7.25E-12
1.00E-05	1.76E-02	9	1.00E-05	0.798	1.000	3.00E-05	6.05E-27	-16.6202	4.94E-12
3.16E-05	8.16E-02	8.9999	3.16E-05	0.798	1.000	9.49E05	1.91E-26	-12.7475	3.36E-12
1.00E04	3.79E-01	8.9999	1.00E-04	0.798	1.000	3.00E-04	6.05E-26	-8.8745	2.29E-12
3.16E-04	1.76E+00	8.9997	3.16E-04	0.797	1.000	9.49E-04	1.91E-25	-5.0006	1.56E-12
1.00E-03	8.80E+00	8.9992	1.00E-03	0.796	1.001	3.00E-03	6.10E-25	-1.1236	7.84E-13
1.10E-03	1.05E+01	8.9992	1.10E-03	0.796	1.001	3.31E-03	6.73E-25	-0.7916	6.28E-13
2.30E-03	1.34E+02	8.9990	6.89E-04	0.795	1.002	5.30E-03	4.21E-25	-0.0002	2.90E-16

Some interesting results are summarized below:

The total molalities of Na and dissolved carbonates and pH remain nearly constant throughout the simulation, whereas the total molalities of dissolved Al and SiO<sub>2</sub> experience remarkable increases from "virtually zero" in the beginning to the final concentrations of 0.689 and 5.30 mmol kg<sup>-1</sup>, respectively. This increment in total Al is accompanied by an increase in the activity of Al ion (of course the initial value of  $6.05 \times 10^{-42}$  is meaningless). Also the thermodynamic affinity changes substantially during the simulation attaining the equilibrium value ("virtually zero") in the last step, after 134 days.

Interestingly, the rate of albite dissolution experiences a substantial decrease (again, the initial value has no meaning) and this change is dictated by the increase in Al ion activity, apart from the decrease in rate for A > -2 kcal mol<sup>-1</sup>. These findings are in marked contrast with the constant rate that would be predicted, at constant pH, by a TST-based rate law, comprising the proton-, water- and hydroxyl-promoted mechanisms, apart from the decrease in rate for A > -2 kcal mol<sup>-1</sup>. These considerations underscore the importance of the approach by Oelkers, Schott and coworkers for a correct evaluation of the dissolution rate of silicate minerals. Unfortunately, this approach cannot be used extensively due to the present lack of experimental data.

The albite surface area specified in the EQ6 input files of Table 7.6,  $A_{s,tot} = sk =$  1,000, represents the area (in cm<sup>2</sup>) in contact with (i.e. exposed to) 1,000 g of water. To understand the physical meaning of this parameter, let us consider an hypothetical system made up of  $N_s$  minerals spheres, all of the same radius  $r_s$ , and 1,000 g of water occupying the pore spaces (porosity is assumed to be 0.3), although natural systems can be different from this simple geometric analogue. The total volume of the mineral spheres,  $V_{tot,s}$ , can be computed based on the definition of porosity,  $\eta$ :

$$\eta = \frac{V_{\rm w}}{V_{\rm w} + V_{\rm tot,s}},\tag{7-44}$$



Figure 7.3. Progressive dissolution of albite in pure water at 25°C, 1 bar, simulated by means of EQ6. (a) Relations between time and reaction progress for two different values of the albite surface area and (b) Inverse correlation between the time needed to attain saturation with albite and the albite surface area.

where  $V_{\rm W}$  identifies the pore volume occupied by water. For a selected radius  $r_{\rm s}$ , the volume of a single mineral sphere  $V_{\rm s}$  is calculated by means of equation (6-93), and the number of minerals spheres is  $N_{\rm s} = V_{\rm tot,s}/V_{\rm s}$ . The surface of a single mineral sphere,  $A_{\rm s}$ , is obtained by using equation (6-92) and the total surface area is  $A_{\rm s,tot} = A_{\rm s}N_{\rm s}$ . These simple calculations indicate that the lowest value of the total surface area, 851 cm<sup>2</sup>, is obtained for  $N_{\rm s} = 1$  ( $N_{\rm s}$  cannot be less then 1!) and  $r_{\rm s} = 8.23$  cm. Both  $A_{\rm s,tot}$  and  $N_{\rm s}$  increase as  $r_{\rm s}$  decreases; for instance, a total surface area of 5,000 cm<sup>2</sup> pertains to a system comprising 203 spheres whose radius is 1.4 cm;  $A_{\rm s,tot}$  becomes 50,000 cm<sup>2</sup> for a system made up of 203,004 spheres whose radius is 0.007 cm.

During the EQ6 simulation, the surface area of albite can be kept constant or automatically changed by the code in proportion to its remaining mass. In the considered example, the difference between these two approaches is negligible, owing to the comparatively high mass of albite present in the system with respect to dissolved mass of albite.

In different EQ6 runs, the surface area of albite was varied from a minimum of  $1,000 \text{ cm}^2$  up to a maximum of  $1,000,000 \text{ cm}^2$ , by changing accordingly the input file of Table 7.6a. It is important to underscore that these differences in the albite surface area do not change the results of the simulation (which are depicted in Fig. 7.2), apart from time. The reason for this is that the number of moles of any species, i.e. solutes, primary phases destroyed and secondary phases produced during the progressive dissolution of albite in pure water at 25°C, 1 bar are uniquely described by the reaction progress variable, as indicated by equation (7-11).

The relations between time and  $\xi$  are shown in Fig. 7.3a for the minimum and maximum values of the albite surface area introduced in the simulation. At the extreme right of this plot, a dashed-vertical line marks albite saturation, which is uniquely defined by  $\xi = 0.01323$  mol. The different times needed to attain this condition for different values of the albite surface area are plotted in Fig. 7.3b as a function of the albite surface area itself. As expected, this plot confirms that there is a simple inverse linear relation between the logarithms of these two variables, namely, log time<sub>sat</sub> (years) =  $-1.00 \times \log A_{s,albite}$  (cm<sup>2</sup>) + 6.00. Therefore, uncertainties in the surface area of reacting minerals brings about corresponding uncertainties in the timing of the dissolution/precipitation process.

## 7.3. Reaction path modelling of geological CO<sub>2</sub> sequestration in ultramafic rocks

As discussed by Lackner et al. (1995, 1997), ultramafic rocks have the highest potential of  $CO_2$  sequestration through mineral fixation, owing to their high concentrations of MgO and CaO. Ultramafic rocks are chiefly made up of four minerals, i.e. olivine, orthopyroxene, clinopyroxene and hornblende, accompanied by small amounts of biotite, garnet and spinel (Le Maitre, 2002). Ultramafic rocks are customarily divided in three broad groups, namely peridotites, pyroxenites and hornblendites, whose names evidently indicate the prevailing mineral. Peridotites contain more than 40% by volume of olivine and, based on the relative amounts of olivine (ol), orthopyroxene (opx) and clinopyroxene (cpx), they are further subdivided in dunite (ol > 90%), harzburgite (ol < 90%,

cpx < 10%), wehrlite (ol < 90%, opx < 10%) and lherzolite (ol < 90%, cpx > 10%, opx > 10 %) (Le Maitre, 2002). Pyroxenites are further subdivided in clinopyroxenite, websterite, orthopyroxenite, olivine-clinopyroxenite, olivine-websterite and olivine-orthopyroxenite (see Le Maitre, 2002).

A dunite contains 49.5 wt% MgO and 0.3 wt% CaO only, a harzburgite is constituted by 45.4 wt% MgO and 0.7 wt% CaO, whereas a lherzolite is made up of 28.1 wt% MgO and 7.3 wt% CaO on average (Lackner et al., 1995). Ultramafic rocks are often affected by serpentinization and the resulting rocks, serpentinites, typically contain 40 wt% MgO and CaO close to 0 (Lackner et al., 1995).

Simple stoichiometric calculations shows that the amounts of ultramafic rocks required to bind chemically 1 kg of  $CO_2$  through reaction with CaO and MgO are 1.8 kg of dunite, 2.0 kg of harzburgite, 2.7 kg of lherzolite and 2.3 kg of sepentinite (Lackner et al., 1995). Hence dunite has the highest potential of  $CO_2$  sequestration among the ultramafic rocks.

Ultramafic rocks typically occur as components of ophiolite complexes, which are interpreted as slices of ancient oceanic crust and upper mantle which were tectonically uplifted and incorporated into the continental crust. Steinmann (1905) was probably the first one who recognized that ophiolites are igneous rocks emplaced on the ocean floor, an interpretation which was enriched by several details in subsequent years. Ultramafic rocks are relatively common at the earth surface and the total amount of MgO stored by these rocks far exceeds the worldwide coal reserves which are currently estimated to be ~ $10^{16}$  kg (Lackner et al., 1995, 1997).

## 7.3.1. Previous works

Xu et al. (2000, 2004) simulated the geological sequestration of  $CO_2$  into different aquifers, including a dunite, by means of TOUGHREACT (Xu and Preuss, 1998). This code can be used to model reactive transport in a non-isothermal multi-phase system. In simple words, reaction path modelling is coupled with solute-transport modelling to simulate the reactive processes occurring into an aquifer. In TOUGHREACT the reaction and transport equations are solved separately. Consequently, TOUGHREACT can be used for reaction path modelling (or batch geochemical modelling to use the words of Xu and coworkers). Adopting this modelling feature, as Xu et al. (2000, 2004) did, TOUGHREACT ACT is equivalent to EQ3/6.

The dunite case is of interest for the present discussion and is briefly presented here. Xu and coworkers considered an initial porosity of 5% for the dunite rock, which was assumed to be made up of forsterite (85.5% by volume) and fayalite (9.5%). Dissolution rates at 25°C were taken equal to  $10^{-13}$  mol m<sup>-2</sup> s<sup>-1</sup> for forsterite and fayalite and precipitation rates at 25°C were fixed at  $10^{-12}$  mol m<sup>-2</sup> s<sup>-1</sup> for all secondary non-carbonate minerals and  $0.6 \times 10^{-8}$  mole m<sup>-2</sup> s<sup>-1</sup> for magnesite and siderite. Apparent activation energy was considered to be 41.87 kJ mol<sup>-1</sup> for magnesite and siderite and 62.76 kJ mol<sup>-1</sup> for all other solid phases. Initial surface areas (in m<sup>2</sup> dm<sup>-3</sup>) were fixed at 8.55 m<sup>2</sup> dm<sup>-3</sup> for forsterite and 0.95 m<sup>2</sup> dm<sup>-3</sup> for fayalite, which means to assume a total surface area of 10 m<sup>2</sup> dm<sup>-3</sup> and to distribute it in proportion to the volume percentages of the two primary phases. Note that this total surface area value corresponds to a surface area of  $2 \times 10^6$  cm<sup>2</sup> exposed to 1,000 g of water, owing to the relatively low porosity of the dunite rock.

In a background run, a dilute aqueous solution (with initial pH of 7 and initial Eh of -100 mV) was equilibrated with forsterite and fayalite at a temperature of 80°C. The aqueous solution thus attained stable pH and Eh values of 10.7 and -520 mV, respectively, through generation of H<sub>2</sub>, owing to the strongly reducing conditions present in the system, and precipitation of magnetite, chrysotile and even elemental Fe. The production of these minerals would decrease porosity to 0.5% in 15,000 years.

In a distinct simulation, a  $CO_2$  pressure of 260 bar was imposed and temperature was maintained at 80°C. This causes a remarkable decrease in pH down to 4.8 and an Eh increase up to 100 mV. Of course, under these more oxidising conditions (with respect to the background run), both H<sub>2</sub> generation and separation of metallic Fe do not occur anymore. The continuous supply of  $CO_2$  to the system activates the dissolution of forsterite and fayalite and, consequently, the precipitation of magnesite and siderite and, to a lesser extent, of talc and amorphous silica as well. After ~1,000 years, ~100 kg of CO<sub>2</sub> are sequestered into 1 m<sup>3</sup> of the considered system (corresponding to approximately 2000 g kg<sup>-1</sup> water) but these processes cause a decrease in porosity to 0.6%. Therefore, it seems likely that both rock alteration and geological CO<sub>2</sub> sequestration end owing to this reduction in porosity.

It must be underscored the prevailing reaction in the model of Xu and coworkers:

$$0.8 \text{ Mg}_2 \text{SiO}_4 + \text{CO}_2 + 0.2 \text{ H}_2 \text{O} \rightarrow \text{MgCO}_3 + 0.2 \text{ Mg}_3 \text{Si}_4 \text{O}_{10}(\text{OH})_2$$
(7-45)

is characterized by a reaction volume for solids (see Section 5.5),  $\Delta V_r$  of 20.2 cm<sup>3</sup>. In contrast, if the dominant reaction would involve (i) a serpentine mineral instead of forsterite and (ii) chalcedony instead of talc, things might be different. This means to take into account, instead of reaction (7-45), the following reaction:

$$0.33 \text{ Mg}_3 \text{Si}_2 \text{O}_5(\text{OH})_4 + \text{CO}_2 \rightarrow \text{MgCO}_3 + 0.67 \text{ SiO}_2 + 0.67 \text{ H}_2 \text{O},$$
 (7-46)

whose  $\Delta V_r$  would be 7.0 cm<sup>3</sup> only or 11.2 cm<sup>3</sup>, if amorphous silica were precipitated instead of chalcedony. Note that both reactions (7-45) and (7-46) involve 1 mol of CO<sub>2</sub> to obtain comparable results (*see* Section 5.5). Therefore, porosity being equal, a rock including a serpentine mineral as primary phase is probably able to sequester more CO<sub>2</sub> than a rock made up of forsteritic olivine.

Based on this expectation, the geological  $CO_2$  sequestration in serpentinitic rocks was investigated by Cipolli et al. (2004), who modelled the irreversible mass exchanges taking place during high-pressure  $CO_2$  injection into a deep aquifer hosted in these kind of rocks by means of the software package EQ3/6 version 7.2b.

Previous researches by Bruni et al. (2002) and Marini and Ottonello (2002) were devoted to understand the origin of the waters circulating in deep aquifers hosted in ultramafic rocks variably affected by serpentinization, which are representative of the initial (prior to  $CO_2$  injection) aqueous solution. These waters have high pH, typically in the 11.3–11.9 range at the surface outlet, and unusual chemistry, with  $Ca^{2+}$  and  $OH^-$  as prevailing cation and anion, respectively. Other characteristics of these waters are: negative Eh (typically -460 to -520 mV), low Mg<sup>2+</sup> concentrations (as low as 0.001 mg kg<sup>-1</sup>), low total dissolved carbonate (generally between 0.5 and 17 mg kg<sup>-1</sup> as HCO<sub>3</sub><sup>-</sup>) and, consequently, very low  $P_{CO_2}$  values, from  $10^{-8}$  to  $10^{-11}$  bar (as already recalled in Section 5.1.4.1). Besides, Ca-OH waters are close to saturation with respect to calcite and brucite, undersaturated with forsterite and enstatite, and strongly oversaturated with respect to antigorite, chrysotile, talc and tremolite.

To understand the origin of Ca-OH waters, the dissolution of a local serpentinite was simulated by means of the EQ3/6 software package by Bruni et al. (2002) and Marini and Ottonello (2002). In this way, it was shown that Ca-OH compositions are produced through prolonged water–rock interaction in a system closed to  $CO_2$ , under strongly reducing conditions, in agreement with previous findings by Pfeifer (1977). This chemical evolution is due to C depletion in a system closed with respect to C sources. This C depletion was attributed to calcite precipitation by Bruni et al. (2002), although the possible reduction of carbonate-C to organic-C and ultimately to  $CH_4$  might also be important. Further research is needed on this subject.

Since serpentinites are almost monomineralic rocks, stoichiometric serpentine  $[Mg_3Si_2O_5(OH)_4]$  was considered to be the only solid phase under dissolution by Cipolli et al. (2004). Simulation was carried out in kinetic mode, assuming constant values for both the dissolution rate of serpentine and its reactive surface area exposed to 1,000 g of water. The latter was set at 4,100 cm<sup>2</sup> based on geochemical evidence. Note that this value is approximately 500 times lower than that considered by Xu and coworkers in their dunite example.

# **7.3.2.** Reaction path modelling of the geological CO<sub>2</sub> sequestration in a serpentinitic rock

#### 7.3.2.1. Setting up the water-rock interaction model

Here, the geochemical model by Cipolli et al. (2004) will be extended to include other two chemical components, Al and Fe. Although serpentinites are almost monomineralic rocks, they main contain significant amounts of Fe oxides ( $7.9 \pm 1.4$  wt% as Fe<sub>2</sub>O<sub>3</sub>, based on the data summarized by Bruni et al., 2002) and Al<sub>2</sub>O<sub>3</sub> ( $2.1 \pm 1.0$  wt%), which follow SiO<sub>2</sub> (40.6  $\pm 1.1$  wt%), MgO ( $35.4 \pm 1.4$  wt%) and H<sub>2</sub>O ( $12.4 \pm 1.6$  wt%), in order of decreasing importance. Assuming that Fe and Al are chiefly present as magnetite and kaolinite, respectively, simple mass balances shows that the molar amounts of serpentine, magnetite and kaolinite for a typical serpentinitic rock (Table 7.7) are 0.9116, 0.0282 and 0.0602, respectively. The corresponding volume fractions are 0.9317, 0.0118 and 0.0565, respectively. Accepting that the total surface area in contact with 1,000 g of water is 4,100 cm<sup>2</sup>, as hypothesized by Cipolli et al. (2004), the initial surface areas of serpentine, magnetite and kaolinite are 3,820, 48.51 and 231.5 cm<sup>2</sup>, respectively. Assuming that initial porosity is 0.3, geometric calculations (*see* Section 7.2.4) indicate that the initial amounts of serpentine, magnetite and kaolinite are and kaolinite in the considered system are 20.04, 0.62 and 1.32 mol, respectively.

Oxide	Wt%		Element	Mol%		
SiO <sub>2</sub>	40.14		Si	10.29		
TiO <sub>2</sub>	0.08		Ti	0.02		
$Al_2 \tilde{O}_3$	2.20	I	Al	0.66		
Fe <sub>2</sub> O <sub>3 tot</sub>	5.88		Fe	1.13		
MnO	0.12		Mn	0.03		
MgO	37.31		Mg	14.25		
CaO	0.18		Ca	0.05		
Na <sub>2</sub> O	0.03		Na	0.01		
K <sub>2</sub> O	0.05		K	0.02		
H <sub>2</sub> O	14.00	ł.	0	49.61		
-			Н	23.93		
Minerals	Mol %	Vol %	Initial moles	Initial surfaces		
Serpentine	91.16	93.17	20.04	3,820.10		
Magnetite	2.82	1.18	0.62	48.51		
Kaolinite	6.02	5.65	1.32	231.54		

Whole-rock chemical analysis of a typical serpentinitic rock from Monte Roccaprebalza (Dinelli et al., 1997) and its computed mineralogic composition

Again, following Cipolli et al. (2004), aquifer temperature was assumed to be 60°C and the aqueous solution hosted in the serpentinitic aquifer was hypothesized to be represented by the local Ca-OH spring water BR2. Heating of this aqueous solution from the emergence temperature, 20.3°C, to the aquifer temperature, 60°C, was simulated by means of EQ6. This process does not cause any change in the total concentrations of solute species, whereas pH decreases from the outlet value, 11.73, to 10.6 and the logarithm of  $f_{O_2}$  increases from -73.6 to -67.9. Results of the heating simulation were used to prepare the EQ3NR input file of Table 7.8. Note that (i) the redox potential is assumed to be fixed by the SO<sub>4</sub><sup>2-</sup>/HS<sup>-</sup> redox couple and (ii) negligible molal concentrations,  $10^{-20}$ , are hypothesized for dissolved Al and Fe, whose analytical concentrations are both below detection limits, to activate these two chemical components.

Injection of pure CO<sub>2</sub> at 100 bar pressure was hypothesized. To this purpose, a constant  $f_{CO_2}$  of 67.9 bar was set in EQ6 input file (*see* Table 7.9). This  $f_{CO_2}$  value is obtained considering that  $\Gamma_{CO_2}$  is equal to 0.679 at 60°C and 100 bar total pressure (*see* Section 3.4). The dissolution of the serpentinitic rock upon continuous CO<sub>2</sub> injection was simulated in kinetic mode, specifying both the dissolution rates at 60°C and the initial surface areas (see above) of the three solid reactants in the EQ6 input file.

The dissolution kinetics of reactant minerals was described by means of the TST rate law, involving the proton- and water-promoted mechanisms, whereas the hydroxyl-promoted dissolution mechanism was neglected, owing to the low pH values attained by the aqueous solution upon  $CO_2$  injection (see below). Owing to the lack of dissolution rate parameters for the proton-promoted mechanism of chrysotile, those of lizardite were assumed to be representative of chrysotile. Dissolution rates at 60°C were computed by

EQ3NR input file to compute the speciation/saturation state of the aqueous solution assumed to be hosted in the serpentinitic reservoir prior to  $CO_2$  injection. This water composition was obtained through heating of spring water BR2 at 60°C (see Cipolli et al., 2004). (Required input formats are not fully respected.)

omissis										
endit.										
tempc = 60	.0000E+0	00								
rho = 1.00000E + 00 fep = 0.00000E + 00 tolbt = 0.00000E + 00			1	tdspkg= 0.	00000E+00	to	spl = 0.00	000E + 00		
			ι	uredox = H						
				toldl = 0.00000E + 00		to	tolsat = 0.00000E + 00			
itermx = 0										
*	1	2	3	4	5	6	7	8	9	10
iopt1-10 =	1	0	0	0	0	0	0	0	0	0
iopg1-10=	0	0	0	0	0	0	0	0	0	0
iopr1-10=	0	0	0	1	0	0	1	-1	0	0
iopr11-20 =	0	0	0	0	0	0	0	0	0	0
iodb1-10=	0	0	0	0	0	0	0	0	0	0
uebal=										
nxmod = 18										
$uxmod = S_{0}$	0									
0	03			-1						
$uxmod = H^{0}$	so <sup>-</sup>			1						
	<b>JO</b> <sub>3</sub>			1						
urmod = SC	(aa)			1						
	<sup>2</sup> (aq)			_1						
urmod - SC	<b>N</b>			-1						
uxiliou – SC	<b>7</b> 3			1						
U unmed — E	0			-1						
$uxmod - S_4$	$\mathbf{U}_{6}$			1						
0	_			-1						
$uxmod = S_5$										
0	_			-1						
uxmod = $S_4$										
0				-1						
uxmod = Hs	$SO_5$									
0	_			-1						
$uxmod = S_2$										
0				-1						
uxmod= CO	D(aq)									
0				-1						
uxmod = Fc	rmaldehy	de(aq)								
0				-1						
uxmod = M	ethanol(a	q)								
0				$^{-1}$						
uxmod= Ao	cetic acid	(aq)								
0				-1						
uxmod= Ad	etate									
0				-1						
uxmod= Fo	rmic acid	l(aq)								
0				-1						

-1 -1

-1

Indee ///
uxmod= Formate
0
uxmod = Ethane(aq)
0
uxmod = Methane(aq)
0
data file master species=H <sup>+</sup>
switch with species =
jflag=16 $csp=-10.6081$
data file master species=Ca <sup>++</sup>
switch with species =
jflag = 3 $csp = 61.8$
data file master species = $Mg^{++}$
switch with species=
jflag = 3 $csp = 0.001$
data file master species = Na <sup>+</sup>
switch with species=
jflag = 3 $csp = 41.1$
data file master species = $K^+$
switch with species=
jflag = 3 $csp = 5.51$
data file master species = $HCO_3^-$
switch with species =
jflag=3 $csp=0.55$
data file master species = $SO_4^{}$
switch with species =
jflag = 3 $csp = 0.3$
data file master species = $Cl^{-}$
switch with species =
jflag = 3 $csp = 30.47$
data file master species = $SiO_2(aq)$
switch with species=
jflag = 3 $csp = 3.1$
data file master species = HS <sup>-</sup>
switch with species=
jflag=3 $csp=0.22$
data file master species = $B(OH)_3(aq)$
switch with species =
jflag = 3 $csp = 0.57$
data file master species = $Fe^{++}$
switch with species=
jflag = 0 $csp = 1.00E-20$
data file master species = $Al^{+++}$
switch with species=
jflag=0 $csp=1.00E-20$
endit.
## TABLE 7.9

EQ6 input file for modeling high-pressure ( $P_{tot} = P_{CO2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer hosted in serpentinitic rocks at 60°C in kinetic mode (time frame). (Required input formats are not fully respected.)

The option switch IOPT1 is set to 1 to direct the code to compute the simulation in time frame The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

endit.										
nmodl1 =	1			nmodl2 = 0	)					
tempc0 = 60.0000E + 00 $tk1 = 0.00000E + 00$				jtemp = 0 tk2 = 0.00000E + 00  tk3 = 0.00000E + 00						
zistrt=	0.00000E + 0	0		zimax = 1	.00000E-0	)6				
tstrt=	0.00000E+0	0		timemx = (	0.00000E +	00				
kstpmx=	500			cplim= (	0.00000E +	00				
dzprnt=	1.00000E+0	0		dzprlg = 0	0.00000E +	00 k	sppmx= 1	00		
dzplot=	0.10000E + 0	0		dzpllg= (	0.00000E +	00 1	ksplmx= 1	0000		
ifile=	60									
*	1	2	3	4	5	6	7	8	9	10
iopt1-10=	1	0	0	0	0	0	0	0	0	0
iopt11-20=	0	0	0	0	0	0	0	0	0	0
iopr1-10=	0	0	0	0	1	0	0	1	0	0
iopr11-20=	0	0	0	0	0	0	0	0	0	0
iodb1-10=	0	0	0	0	0	0	0	0	0	0
iodb11-20=	0	0	0	0	0	0	0	0	0	0
*										
nxopt=	1									
option=	all									
nxopex=	15									
exception=	Gibbsite									
exception=	Kaolinite									
exception=	Chrysotile									
exception=	Sepiolite									
exception=	Hydromagne	esite								
exception=	Nesquehonit	e								
exception=	Magnesite									
exception=	Calcite									
exception=	Dolomite									
exception=	Chalcedony									
exception=	Magnetite									
exception=	Siderite									
exception=	Pvrite									
exception=	Goethite									
exception=	Dawsonite									
*					<b>.</b>					
nffg=	1									
species=	$CO_{1}(g)$									
moffg=	= 1 00000E+(	)1		xlkffg = -	$\pm 1.83187E$	E+00				
*		· •								
nrct=	3									
* reactant=	Chrysotile									
jcode=	0			jreac= (	)					
-				-						

TABLE 7.9 (Continued)

TIBEE 7.9 (Communed)		
morr= 20.0400E+00	modr = 0.00000E + 00	
nsk = 1	sk = 3,820.000E + 00	fk = 0.00000E + 00
nrk = 2	nrpk = 0	
imech = 2		
rk0= 4.89000E-09	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H^+$	cdac = 0.80000E + 00	
rk0= 2.25000E-15	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H_2O$	cdac = 1.00000E + 00	
reactant= Magnetite	·	,,,,,,,,,,,
jcode = 0	jreac = 0	
morr = 0.62000E + 00	modr = 0.00000E + 00	
nsk = 1	sk = 48.5000E + 00	fk = 0.00000E + 00
nrk = 2	nrpk = 0	
imech=2		
rk0 = 5.65000E - 13	trk0 = 5.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H^+$	cdac = 0.27900E + 00	
rk0 = 3.65000E - 15	trk0 = 5.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H_2O$	cdac = 1.00000E + 00	
reactant= Kaolinite		
jcode = 0	jreac = 0	
morr = 1.32000E + 00	modr = 0.00000E + 00	
nsk = 1	sk = 231.500E + 00	fk = 0.00000E + 00
nrk = 2	nrpk = 0	
imech = 2		
rk0 = 8.00000E - 15	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H^+$	cdac = 0.77700E + 00	
rk0 = 1.69000E - 17	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H_2O$	cdac = 1.00000E + 00	
dlzidp= 0.00000E+00		
tolbt = 0.00000E + 00	toldl = 0.00000E + 00	tolx = 0.00000E + 00
tolsat = 0.00000E + 00	tolsst = 0.00000E + 00	
screw1 = 0.00000E + 00	screw2 = 0.00000E + 00	screw3 = 0.00000E + 00
screw4 = 0.00000E + 00	screw5 = 0.00000E + 00	screw6 = 0.00000E + 00
zklogu= 0.000	zklogl = 0.000	zkfac= 0.000
dlzmx1 = 0.00000E + 00	dlzmx2 = 0.00000E + 00	nordlm = 0

 TABLE 7.9 (Continued)

TABLE 7.9 (Continued)		
itermx = 0	ntrymx = 0	
npslmx = 0	nsslmx = 0	ioscan = 0
*		
* pickup file written by EQ3NR, version	n 7.2b (R139 )	
* supported by EQLIB, version 7.2b (R	168)	
Spring water BR2 heated at 60°C		
endit.		
tempci = 6.00000E + 01		
nxmod = 18		
species = $S_2 O_3^{}$		
type = 0	option = -1	xlkmod = 0.00000E + 00
species = $HSO_3^-$		
type = 0	option = -1	x1kmod = 0.00000E + 00
$species = SO_2(aq)$		U
type = 0	option = -1	xikmod = 0.00000E + 00
species = $SO_3$		$r_{\rm ibmod} = 0.00000E \pm 00$
type = 0	option = -1	$XIKIIIOd = 0.00000E \pm 00$
species $-S_4O_6$	ontion = -1	$x$ 1kmod = 0.00000E $\pm$ 00
species $= S^{}$	option – 1	
species $= 3_5$ type = 0	option = $-1$	$x kmod = 0.00000E \pm 00$
species = $S_{}^{}$	option 1	AIRING 0.000001+00
$S_4$ type= 0	option = $-1$	x kmod = 0.00000E + 00
$species = HSO_s^{-1}$	- F	
type = 0	option $= -1$	xlkmod = 0.00000E + 00
species = $S_2^{}$	*	
type = 0	option $= -1$	xlkmod = 0.00000E + 00
species = CO(aq)		
type = 0	option $= -1$	xlkmod = 0.00000E + 00
species = Formaldehyde(aq)		
type = 0	option $= -1$	xlkmod = 0.00000E + 00
species = Methanol(aq)		
type = 0	option $= -1$	xlkmod = 0.00000E + 00
species = Acetic acid(aq)		
type = 0	option = $-1$	xlkmod = 0.00000E + 00
species=Acetate	1	-1000000000000000000000000000000000000
type = 0	option = -1	$xikinod = 0.00000E \pm 00$
tupe= 0	ontion = -1	xlkmod= 0.00000E+00
species = Formate	option- i	
species = 10 $type = 0$	ontion = -1	x kmod = 0.00000 E + 00
species = Fthane(aq)	option 1	
type=0	option = -1	xlkmod = 0.00000E+00
species=Methane(aq)	option 1	
type = 0	option = $-1$	xlkmod = 0.00000E + 00
iong 1 = 0	iopg2=0	iopg3 = 0
iopg4 = 0	iong5=0	iopg6=0
iopg7 = 0	iopg8 = 0	iopg9=0
iopg10=0	10 -	10
kct = 13	ksq = 14	kmt = 14
kxt= 14	kdim= 14	kprs= 0

TABLE 7.9 (	(Continued)
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0	5.551282557733262E+01	
Al	1.00000000335209E-20	
В	9.218375990262418E-06	
Ca	1.541993186674240E-03	
Cl	8.594550329665458E-04	
Fe	1.00000000144592E20	
Н	1.110211073777142E+02	
С	9.013861125756066E-06	
K	1.409268500602602E-04	
Mg	4.114379952680602E-08	
Na	1.787751767510004E03	
S	9.774694209498928E-06	
Si	5.159417928970046E05	
electr	-1.150065192877400E-04	
$H_2O$	H <sub>2</sub> O	1.744358983526984E+00
Al+++	Al <sup>+++</sup>	-4.257641064725003E+01
B(OH) <sub>3</sub> (aq)	$B(OH)_3(aq)$	-6.725193420264618E+00
Ca <sup>++</sup>	Ca <sup>++</sup>	-2.815981346143646E+00
$Cl^-$	$Cl^-$	-3.066044127812017E+00
Fe <sup>++</sup>	Fe <sup>++</sup>	- 2.130818472540851E+01
$H^+$	$H^+$	- 1.057158162952603E+01
$HCO_3^-$	$HCO_3^-$	-6.323495220189614E+00
$\mathbf{K}^+$	$\mathbf{K}^+$	-3.851088497828405E+00
Mg <sup>++</sup>	Mg <sup>++</sup>	-7.387095228892170E+00
Na <sup>+</sup>	Na <sup>+</sup>	-2.748464060961117E+00
$SO_4^{}$	$SO_4^{}$	-5.564691145737220E+00
SiO <sub>2</sub> (aq)	SiO <sub>2</sub> (aq)	-5.509600627218712E+00
O <sub>2</sub> (g)	O <sub>2</sub> (g)	-6.354977441383422E+01

means of equation (6-26), based on the pre-exponential factors,  $A_i$ , and apparent activation energies,  $E_i$ , given in Table 6.1. Again, note that the reaction order of the water-promoted mechanism was set to +1.000 in the EQ6 input file (Table 7.9). The EQ6 code was instructed to change the surface areas of primary minerals during the simulation in proportion to their remaining masses.

Instantaneous partial equilibrium was assumed for precipitating, secondary solid phases. Again, this choice is due to the difficulty in defining the surface area for minerals which are initially absent in the considered system. The hypothesis of instantaneous equilibrium for secondary minerals means to assume that the dissolution of reactants is the rate limiting step of the overall process. Results are presented against both the reaction progress variable,  $\xi$ , and time in the plots of Figs. 7.4 to 7.7.

## 7.3.2.2. Solid reactants

The amounts of reactant minerals destroyed during continuous  $CO_2$  injection are shown as a function of time and reaction progress in Fig. 7.4. The solid phase contributing most chemical components to the considered system through its destruction is chrysotile. This reflects not only its high abundance and surface area (see Table 7.7) but



Figure 7.4. Moles of reactant minerals destroyed during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks as a function of time and the reaction progress variable.

also the comparatively high dissolution rate (in mol  $cm^{-2} s^{-1}$ ) of serpentine minerals in acidic solutions (e.g., lizardite in Table 6.1, since the dissolution rate of chrysotile in acidic solutions was never investigated, as already recalled).

Although kaolinite is much more abundant than magnetite (see Table 7.7), the moles of magnetite destroyed are over two orders of magnitude higher than those of kaolinite, owing to the higher dissolution rate (in mol cm<sup>-2</sup> s<sup>-1</sup>) of magnetite with respect to kaolinite (see Table 6.1).

The aqueous solution remains strongly undersaturated with respect to chrysotile  $(A \le -16 \text{ kcal mol}^{-1})$  and magnetite  $(A \le -3.4 \text{ kcal mol}^{-1})$  throughout the process. For these low values of *A*, the dissolution rate is virtually independent of thermodynamic affinity (see Section 6.4.2) and is a function of pH only.

In contrast, the aqueous solution saturates with kaolinite for  $\xi$  of 0.025 mol. Before attainment of this condition the thermodynamic affinity with respect to kaolinite increases significantly, determining a substantial decrease in the dissolution rate. For instance, the dissolution rate of kaolinite is  $1.90 \times 10^{-17}$  mol cm<sup>-2</sup> s<sup>-1</sup> at  $\xi$  of 3.16 mmol, where *A* is -3.59 kcal mol<sup>-1</sup> and pH is 4.49, but it reduces to  $3.00 \times 10^{-18}$  mol cm<sup>-2</sup> s<sup>-1</sup> at  $\xi$  of 10 mmol, where *A* is -0.12 kcal mol<sup>-1</sup> and pH is 4.85. Upon attainment of equilibrium with kaolinite, this phase was dropped from the considered primary paragenesis.



Figure 7.5. Moles of product minerals formed during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks as a function of time and the reaction progress variable. Note that letters A to E mark the onset of the precipitation of different secondary minerals.

Chrysotile is exhausted after 4,300 years, for  $\xi$  of 20.04 mol, which corresponds to the initial amount of this mineral in the considered serpentinitic rock. The simulation ends for  $\xi$  of 20.66 mol, upon exhaustion of magnetite.

## 7.3.2.3. Solid product phases

The solid phases which are considered to be produced through high-pressure ( $P_{tot} = P_{CO_2} = 100 \text{ bar}$ ) CO<sub>2</sub> injection in a deep aquifer hosted in serpentinitic rocks are, in order of appearance, chalcedony, pyrite, dolomite, magnesite, goethite and siderite (Fig. 7.5). However, pyrite is produced in very small amounts ( $< 3 \mu$ mol) and has ephemeral existence; dolomite never exceeds 0.001 mol; goethite and siderite are always  $\leq 0.1 \text{ mol}$ . As already recognized by Cipolli et al. (2004), only chalcedony and magnesite are produced in significant quantities and for  $\xi > 0.05$  mol the stoichiometry of the process is adequately described by reaction (7-46). In other words, 3 mol of magnesite and 2 mol of chalcedony are produced for each mole of serpentine dissolved. The masses of magnesite and chalcedony increase almost linearly with time (not shown) and the magnesite mass is 2.32 mol kg<sup>-1</sup> water (196 g) after 10 years and 10.4 mol kg<sup>-1</sup> water (874 g) after 50 years, and so on, indicating the high CO<sub>2</sub> sequestration capacity of this process. A time of tens of years is apparently large, but it is actually small if compared with the residence times of high-pH Ca-OH waters in deep aquifers, which is estimated to be in the order of 100 to 10,000 years (Cipolli et al., 2004).



Figure 7.6. Concentrations of solutes during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks as a function of time and the reaction progress variable. Note that letters A to E marks the onset of the precipitation of different secondary minerals as shown in Figure 7.5.

## 7.3.2.4. The aqueous solution

Throughout continuous CO<sub>2</sub> injection, the main solute is aqueous CO<sub>2</sub>, whose concentration is fixed at 1.1 mol kg<sup>-1</sup> (48,400 ppm) by constant  $f_{CO_2}$ , under specified temperature-pressure conditions (Fig. 7.6). Aqueous CO<sub>2</sub> is followed by HCO<sub>3</sub><sup>-</sup> and Mg<sup>2+</sup>, whose concentrations, after an initial increase (for  $\xi < 0.01$ ), stabilize at 0.062 and 0.028 mol kg<sup>-1</sup>, respectively, when magnesite begins to form at point C. Note that at this point also Ca<sup>2+</sup>concentration is fixed by saturation with dolomite which is attained at point B.

Also plotted in Fig. 7.6 is chloride, the most typical mobile solute, whose concentration experiences a significant decrease for high values of  $\xi$ . This effect is caused by a substantial increase in the mass of water, which is provided to the considered system by the dissolution of chrysotile, a H<sub>2</sub>O-containing mineral. This effect is exactly the opposite of dry-up, a phenomenon due to incorporation of water in hydrous minerals (see Reed, 1997).

During the considered process, the pH of the aqueous solution changes from 3.83 to 4.85 (Fig. 7.7a). Most of this pH increase occurs between the onset of chalcedony precipitation (point A) and the beginning of magnesite precipitation (point C), after which pH is buffered by carbonate equilibria, similar to what was discussed for the aqueous solution/calcite example in Section 7.1.1.



Figure 7.7. Evolution of (a) pH and (b)  $f_{0,2}$  during high-pressure ( $P_{tot} = P_{C0,2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks as a function of time and the reaction progress variable. Note that letters A to E mark the onset of the precipitation of different secondary minerals as shown in Figure 7.5.

#### Reaction Path Modelling of Geological CO<sub>2</sub> Sequestration

In spite of its relatively low concentration in the aqueous phase and the small amounts of Fe-bearing secondary minerals produced during continuous CO<sub>2</sub> injection, equilibria between Fe product minerals and the aqueous solution govern the  $f_{O_2}$  of the system, at least after the onset of goethite precipitation (Fig. 7.7b, point D). This event brings about a sharp, remarkable increase in the logarithm of oxygen fugacity from -59.3 to -42.4. The log  $f_{O_2}$ is finally fixed at -47.53, for  $\xi > 2.7$  (to the right of point E), by coexistence of goethite and siderite at constant  $f_{CO_2}$ , temperature and pressure, as expressed by the reaction:

$$FeOOH_{(s)} + CO_{2(g)} = FeCO_{3(s)} + \frac{1}{4}O_{2(g)} + \frac{1}{2}H_2O.$$
 (7-47)

#### 7.3.2.5. The $CO_2$ sequestration

The masses of  $CO_2$  sequestered during the considered process through dissolution (and speciation) in the aqueous solution (solubility trapping) and through incorporation in solid carbonates, mainly magnesite and subordinately siderite and dolomite (mineral fixation) are plotted against both the reaction progress and time in Fig. 7.8. The total mass of sequestered  $CO_2$ , corresponding to the sum of these two terms, is also shown.



Figure 7.8. High-pressure ( $P_{tot} = P_{CO_1} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks: masses of CO<sub>2</sub> sequestered through dissolution in the aqueous solution (solubility trapping) and incorporation in solid carbonates (mineral fixation), and total mass of sequestered CO<sub>2</sub> as a function of time and the reaction progress variable. Note that letters A to E mark the onset of the precipitation of different secondary minerals as shown in Figure 7.5.

Inspection of Fig. 7.8 shows that CO<sub>2</sub> sequestration through dissolution in the aqueous solution is instantaneous, but relatively limited, as the maximum amount of CO<sub>2</sub> which can be dissolved in 1 kg of water is ~50 g, at the specified conditions of 100 bar  $P_{tot} = P_{CO_2}$ and 60° C. Carbon dioxide incorporation in secondary solid carbonates begins later on, namely at points B (dolomite), C (magnesite) and E (siderite). The log-mass of CO<sub>2</sub> incorporated in precipitating solid carbonates increases linearly with log  $\xi$ , attaining values much greater than the mass of CO<sub>2</sub> dissolved in the aqueous solution, but this process requires comparatively long time intervals. For instance, the contribution of mineral fixation attains the same level of solubility trapping after 5 years. However, mineral carbonation becomes more and more important afterwards, whereas solubility trapping does not change with time. Again, the sequestration capacity of the process is large and time is less than the residence times of high-pH waters in deep aquifers. In particular the value of 2,000 g kg<sup>-1</sup> water is attained in 350 years. The same sequestration capacity is attained in 1,000 years in the dunite model by Xu and coworkers, who used somewhat different figures for several parameters (*see* Section 7.3.1).

## 7.3.2.6. Changes in the porosity of aquifer rocks

To evaluate the changes in the porosity of aquifer rocks in response to high-pressure ( $P_{tot} = P_{CO2} = 100$  bar) CO<sub>2</sub> injection we continue to make reference to 1 kg of water. In this way, the model is kept completely independent of the effective porosity of the aquifer. For our purpose, it is useful to take into account the percentual change in the volume of solid phases,  $\Delta V/V$ . This parameter was already defined in equation (5-125). Here it is re-defined, more correctly, as follows:

$$\Delta V/V = \frac{\sum_{P} n_{P} \cdot v_{P}^{\circ} - \sum_{R} n_{R} \cdot v_{R}^{\circ}}{\sum_{R} n_{R} \cdot v_{R}^{\circ}} \cdot 100$$
(7-48)

where  $v_i^{o}$  and  $n_i$  represent the molar volume and the moles of the *i*th solid phase, respectively, and subscripts *P* and *R* indicate products and reactants, respectively. Indeed equation (7-48) is similar to the simplified relation (5-125), in which the stoichiometric coefficients of solid phases were used instead of their moles.

The variation of the  $\Delta V/V$  parameter as a function of both reaction progress and time is shown in Fig. 7.9. During the first stages of the process (for  $\xi < 0.0003$  moles) the  $\Delta V/V$  is -100% as rock-forming minerals dissolve congruently, i.e. without precipitation of secondary solid phases. Formation of chalcedony (point A) brings about an initially sharp increase in  $\Delta V/V$  and its subsequent stabilization close to -60%. Upon the onset of carbonate minerals precipitation, dolomite at point B and magnesite at point C, the  $\Delta V/V$ parameter experiences a new initially sharp increase and gradually approaches +19.3%, which is the limiting value constrained by dissolution of chrysotile accompanied by stoichiometric precipitation of magnesite and quartz (or chalcedony; note that the molar volumes of chalcedony and quartz are virtually equal; see Table 5.11). A value of  $\Delta V/V$  close to 17% is attained in ~4 years (Fig. 7.9).



Figure 7.9. Variation of the  $\Delta V/V$  parameter as a function of both reaction progress and time during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in serpentinitic rocks. Note that letters A to E mark the onset of the precipitation of different secondary minerals as shown in Figure 7.5.

Obviously, for  $\Delta V/V = 0$ , the rock-forming minerals dissolved (mainly serpentine) are substituted by an equal volume of secondary solid phases and the porosity of the system does not change. Unfortunately, in the case under consideration, the  $\Delta V/V$  is positive and significantly different from zero. Consequently, the occurring reactions, mainly serpentine dissolution accompanied by precipitation of magnesite and chalcedony causes a progressive reduction in the effective porosity of the aquifer. This would drop to zero if its initial value were 19%, whereas effective porosities >0 are only possible for initial values >19%. If precipitation of amorphous silica takes place instead of chalcedony, the reduction in porosity could be even larger, with a maximum  $\Delta V/V$  of 30.9%.

Although an initial porosity of 19% seems to be too high a value for serpentinites, it must be emphasized that (1) the natural system might behave as an open, flow-through system (rather than as a closed system), in which case secondary solid phases might deposit outside the reference volume affected by serpentine dissolution; (2) fracturing induced by injection of pressurized CO<sub>2</sub> might occur, with consequent increases in the effective porosity and permeability. As a matter of fact, magnesite veins with nodules of opal are common in the serpentinitic rocks of Southern Tuscany (Arisi Rota et al., 1971), a region characterized by high thermal fluxes (>100 mW m<sup>-2</sup>, Baldi et al., 1994) and high

 $CO_2$  fluxes (Marini and Chiodini, 1994), indicating that reaction (7-46) occurred naturally in these environments.

In contrast, as the magnesite and silica precipitate in the serpentinitic aquifer, these minerals will likely armor the remaining reactants, thereby further decreasing effective surface area in addition to decreasing porosity. These effects could represent serious obstacles for the implementation of this methodology of  $CO_2$  sequestration and their importance must be evaluated by means of laboratory experiments first and field tests afterwards.

# **7.3.3. Reaction path modelling of the geological CO<sub>2</sub> sequestration in a serpentinitic aquifer: salinity effects**

## 7.3.3.1. Setting up a water-rock interaction model involving a brine

To investigate the influence of salinity on the geological sequestration of  $CO_2$ , it would be desirable to carry out the same simulation of Section 7.3.2, but involving a brine instead of a dilute aqueous solution. As discussed in Sections 4.4 and 4.5, the best way to take into account the interactions among solutes in very concentrated electrolytes is by use of the Pitzer's equations. This capability is available in the software package EQ3/6 and we took advantage of it to carry out some calculations in Section 4.5. Unfortunately, the Pitzer's interaction parameters are not available for fundamental chemical components such as SiO<sub>2(aq)</sub> and Al<sup>3+</sup> and are also restricted at relatively low temperatures, sometimes at 25°C only. This is the case, for instance, of the model by Harvie, Møller and Weare (1984) and the corresponding HMW database of EQ3/6, which is based on the system Na-K-Mg-Ca-H-Cl-SO<sub>4</sub>-OH-HCO<sub>3</sub>-CO<sub>3</sub>-CO<sub>2</sub>-H<sub>2</sub>O and works at 25°C only, as already recalled in Section 4.4.2.

If we want to perform the simulation of Section 7.3.2 but taking into account a brine, we have to accept some limitations and introduce some assumptions. One possibility, that is further detailed below, is to carry out a simplified simulation at the temperature of 25°C, considering only chrysotile as reactant. Accepting to work at 25°C, the HMW database can be used, but  $SiO_{2(aq)}$  has to be introduced into it. Strictly speaking, this requires the time-consuming search and processing of the experimental data (admitting that they do exist !) which are needed to derive the Pitzer's interaction parameters between  $SiO_{2(aq)}$  and other chemical species, an exercise that goes well beyond the purpose of this book. This serious obstacle can be overcome through a sort of crafty trick, that is hypothesizing that the interaction parameters of  $SiO_{2(aq)}$  are equal to those of dissolved  $CO_2$ , since both species are neutral. The HMW database was therefore changed accordingly. Of course, this is a very rough choice, but still it is better than nothing.

Again, we assume to inject pure CO<sub>2</sub> at a fugacity of 67.9 bar. Note that since at the considered temperature of 25°C, the fugacity coefficients of CO<sub>2</sub> are smaller than at 60°C (*see* Figure 3.5b), a total pressure (= $P_{CO_2}$ ) of 263 bar (where  $\Gamma_{CO_2}$  is 0.258) is needed to attain this  $f_{CO_2}$  value.

Two simulations were carried out, one involving a brine, initially at saturation with halite, the other with pure water. The results of the pure-water run represent the term of comparison to evaluate salinity effects. Indeed, the simulation of Section 7.3.2 cannot be used for reference due to differences not only in salinity but also in temperature, pressure,

primary phases and secondary minerals (these are really too many differences). The EQ3 input file for the brine was prepared setting a pH of 7, imposing halite saturation to constrain the concentration of Na<sup>+</sup>, and assuming  $m_{\text{Na}^+,\text{tot}} = m_{\text{Cl}^-,\text{tot}}$  Negligible total concentrations ( $10^{-20}$  mol kg<sup>-1</sup>) were assigned to Mg<sup>2+</sup>, HCO<sub>3</sub><sup>-</sup> and SiO<sub>2(aq)</sub> to activate these chemical components. Pure water was assumed to have initial pH 7 and initial negligible concentrations ( $10^{-20}$  mol kg<sup>-1</sup>) of Mg<sup>2+</sup>, HCO<sub>3</sub><sup>-</sup> and SiO<sub>2(aq)</sub>.

As anticipated, chrysotile was considered to be the only primary solid phase. Values of 4,100 cm<sup>2</sup> and 20.04 mol were assigned to its initial surface area exposed to 1,000 g of water and to its initial amount (see Section 7.3.2.1). Dissolution rate constants were set at  $2.00 \times 10^{-10}$  and  $1.00 \times 10^{-16}$  mol cm<sup>-2</sup> s<sup>-1</sup> for the proton- and water-promoted mechanisms, respectively. These are the values for lizardite and chrysotile, respectively, listed in Table 6.1. The reaction orders with respect to H<sup>+</sup> and H<sub>2</sub>O are 0.8 and 1.0, respectively, as reported in Table 7.9. Chalcedony, magnesite and halite (in the brine run) were considered to be the only precipitating solid-product phases and instantaneous partial equilibrium was assumed for all secondary minerals.

### 7.3.3.2. Solid-product phases

The amounts of secondary solid phases produced through high-pressure ( $P_{tot}$  =  $P_{\rm CO_2} = 263$  bar) CO<sub>2</sub> injection in a system comprising chrysotile and an aqueous solution are displayed in Figure 7.10 for the two runs, involving either a brine, initially saturated with halite, or pure water. In this way the two sets of results can be compared and possible salinity effects can be evaluated. In the pure water run, the onset of chalcedony and magnesite precipitation is somewhat delayed with respect to the brine run, as indicated by the spacings between points A and A' and between points B and B'. However, the amounts of chalcedony and magnesite produced are equal for  $\xi > 0.003$  and  $\xi > 0.1$  mol, respectively. The most striking difference between the two simulations, however, is the appearance, in the brine run, of halite whose precipitation occurs immediately upon CO<sub>2</sub> injection. Since the brine was initially saturated with halite under a negligible  $f_{CO_2}$ , close to  $10^{-20}$  bar, the increase in  $f_{\rm CO_3}$  makes the brine to become oversaturated with halite and a significant amount of this mineral, 0.24 moles, is immediately precipitated. Upon progressive chrysotile dissolution (for  $\xi < 0.1$  mol approximately), no further precipitation of halite occurs and this mineral simply remains in the system. However, the dissolution of chrysotile also causes an increase in the mass of water (as already recalled in Section 7.3.2.4, see Figure 7.11a), which becomes significant for  $\xi > 0.1$  mol approximately. This increase in water brings about re-dissolution of halite, which disappears from the system at  $\xi = 1$  mol (point C). For higher values of the reaction progress, the two runs provide the same results in terms of precipitating solid phases.

#### 7.3.3.3. The aqueous solution

The concentrations of relevant solutes and pH during progressive chrysotile dissolution triggered by high-pressure ( $P_{tot} = P_{CO_2} = 263$  bar) CO<sub>2</sub> injection are shown in Figure 7.11 for both the brine run and the pure-water run, to facilitate the comparison of the two series of results and to infer salinity effects.



Figure 7.10. Moles of product minerals formed during high-pressure ( $P_{tot} = P_{CO_2} = 263$  bar) CO<sub>2</sub> injection in a system ( $T = 25^{\circ}$ C) comprising chrysotile and an aqueous solution, either a brine, initially halite-saturated (solid black lines and black time scale) or pure water (dashed grey lines and grey time scale), as a function of time and the reaction progress variable.

First, it must be underscored that after halite re-dissolution (point C), the addition of water to the system through chrysotile dissolution continues and brings about a significant decrease in the concentrations of  $Na^+$  and  $Cl^-$ , which behave as mobile components in the final part of the brine simulation.

The main differences between the two simulations, especially before halite re-dissolution, are in pH and in the concentrations of  $CO_{2(aq)}$  and  $SiO_{2(aq)}$  (although the latter ones can be biased by the assumption on the interaction parameters between  $SiO_{2(aq)}$  and the other dissolved species), whereas the differences in the concentrations of  $Mg^{2+}$  and  $HCO_{3}^{-}$  are comparatively small.

Dissolved CO<sub>2</sub> concentration in the halite-saturated brine is 32.3 g kg<sup>-1</sup>, approximately 1/3 of that in pure water, 101 g kg<sup>-1</sup>. The CO<sub>2(aq)</sub> concentration increases somewhat upon halite re-dissolution and brine dilution, i.e. after point C, and attain the maximum value of 48.8 g kg<sup>-1</sup>, about  $\frac{1}{2}$  of the pure water value, at the end of the simulation.

The pH of the halite-saturated brine increases from the initial value of 2.27 to 4.48 at point C but remains 0.7 to 0.5 pH-units lower than the pure-water pH. The maximum



Figure 7.11. (a) concentrations of relevant solutes, mass of water, and (b) pH during high-pressure ( $P_{tot} = P_{CO_2} = 263$  bar) CO<sub>2</sub> injection in a system ( $T = 25^{\circ}$ C) comprising chrysotile and an aqueous solution, either a brine, initially halite-saturated (solid black lines and black time scale) or pure water (dashed grey lines and grey time scale), as a function of time and the reaction progress variable. Letters A, A', B, B' and C correspond to those in Figure 7.10.

increase in pH takes place between the beginning of chalcedony precipitation (point A, pH = 2.33) and the onset of magnesite precipitation (point B, pH = 4.48), after which pH is fixed by carbonate equilibria, as already noticed in Sections 7.1.1 and 7.3.2.4. The difference in pH between the brine and pure water reduces after point C due to dilution of the brine.

## 7.3.3.4. The CO<sub>2</sub> sequestration

The total mass of CO<sub>2</sub> sequestered through high-pressure ( $P_{tot} = P_{CO_2} = 263$  bar) CO<sub>2</sub> injection in a system made up of chrysotile and a brine, initially saturated with halite is compared with that sequestered in a reference system constituted by chrysotile and pure water in Figure 7.12. Also shown are the separated contributions of CO<sub>2</sub> dissolution (and speciation) in the aqueous solution (solubility trapping) and incorporation in precipitating magnesite (mineral fixation).

The plot shows that solubility trapping is strongly dependent on the salinity of the aqueous solution. Nevertheless, the spacing between the two lines outlining solubility trapping, for the brine and pure water, decreases with increasing reaction progress owing



Figure 7.12. High-pressure ( $P_{tot} = P_{CO_1} = 263$  bar) CO<sub>2</sub> injection in a system ( $T = 25^{\circ}$ C) comprising chrysotile and an aqueous solution, either a brine, initially halite-saturated (solid black lines and black time scale) or pure water (dashed grey lines and grey time scale). The masses of CO<sub>2</sub> sequestered through both dissolution in the aqueous solution (solubility trapping) and incorporation in precipitating magnesite (mineral fixation) as well as the total mass of sequestered CO<sub>2</sub> are shown as a function of time and the reaction progress variable. Letters A, A', B, B' and C correspond to those in Figure 7.10.

to progressive dilution of the brine, as discussed in the previous section. In contrast, mineral fixation is similar for the chrysotile–brine and the chrysotile–pure water systems. Consequently, the difference between the total masses of  $CO_2$  undergoing geological sequestration in these two distinct systems decreases with increasing reaction progress owing to the decreasing contribution of solubility trapping and the increasing contribution of mineral fixation.

Finally, it must be underscored that the time scales are different for the two considered systems. The time needed to exhaust 95% of chrysotile is 12,650 years for the chrysotile–brine system, whereas it amounts to 23,080 years for the chrysotile–pure water system. These values are much higher than the corresponding 758 years of the simulation presented in Section 7.3.2, owing to the different temperatures, 60°C in Section 7.3.2 against 25°C in this section.

# 7.4. Reaction path modelling of geological CO<sub>2</sub> sequestration in continental tholeiitic flood basalts

Continental tholeiitic flood basalts represent another type of rocks with a relatively high potential of  $CO_2$  sequestration, owing to relatively high contents of MgO and CaO, namely 6.2 and 9.4 wt% on average, respectively (Lackner et al., 1995, 1997). The  $CO_2$  sequestration potential is enhanced further by the significant Na<sub>2</sub>O and FeO contents, 2.7 and 9.2 wt% on average, respectively, based on the data reported by Wilson (1989).

A typical example of this kind of rocks is represented by the Columbia River tholeiitic flood basalts in northwestern USA, which extend over an area of ~200,000 km<sup>2</sup> with an average thickness of ~1 km and a maximum thickness> 1.5 km (Wilson, 1989). This volume of rocks would be enough to fix over twice the amount of CO<sub>2</sub> produced through combustion of the global coal reserves, which are estimated to be in the order of  $10^{16}$  kg (Lackner et al., 1995), as already recalled above.

Among the other major continental flood-basalt provinces, we may quote the Siberian Platform, with an extension  $> 1,500,000 \text{ km}^2$  and a maximum thickness of 3.5 km and the Paranà (Brazil)-Etendeka (Namibia) province, covering a total area of 1,200,000 km<sup>2</sup> with a maximum thickness of 1.8 km (Wilson, 1989).

Therefore it is instructive to model the sequestration of  $CO_2$  through reaction with this type of rock.

## 7.4.1. Setting up the water–rock interaction model

Starting from the whole-rock chemical analysis of a continental tholeiitic flood basalt from the Columbia River Province (Table 7.10) and assuming that its main constituting mineral phases are a labradoritic plagioclase ( $X_{\text{Anorthite}} = 0.55$ ,  $X_{\text{Albite}} = 0.45$ ), clinopyroxene ( $X_{\text{Diopside}} = 0.75$ ,  $X_{\text{Hcdenbergite}} = 0.25$ ), orthopyroxene ( $X_{\text{Enstatite}} = 0.60$ ,  $X_{\text{Ferrosilite}} = 0.40$ ), magnetite, apatite and ilmenite (as suggested by available petrographic data, see Wilson, 1989), simple mass balance calculations show that the molar fractions of these phases are 0.5454, 0.1968, 0.1464, 0.0253, 0.0047 and 0.0210, respectively. Apatite and ilmenite were neglected from subsequent data elaboration owing to the low content and relatively high chemical durability, respectively; rock composition was then recomputed imposing that the sum of the molar fractions of the other minerals is equal to 1. In this way, the mineralogic composition reported in Table 7.10 was obtained.

The total surface area in contact with 1,000 g of water was set again at 4,100 cm<sup>2</sup>, as in the serpentinite case (Sections 7.3.2 and 7.3.3), for comparative purposes. It was distributed in proportion to the volume percentages of the main rock-forming minerals obtaining the initial surface areas reported in Table 7.10. Their initial amounts in the considered system (which are also given in Table 7.10) were then computed through simple geometric computations (*see* Section 7.2.4), assuming an initial porosity of 0.3.

Again, for comparative purposes, aquifer temperature was kept at 60°C and injection of pure CO<sub>2</sub> at a constant pressure of 100 bar ( $f_{CO_2}$  of 67.9 bar) was imposed. The initial aqueous solution was assumed to be in equilibrium with analcime, calcite, kaolinite, daphnite, clinochlore, chalcedony and magnetite at 60°C, 1 bar under a  $f_{CO_2}$  of 10<sup>-1</sup> bar (*see* Section 7.1.2).

The computer simulation was performed in kinetic mode, specifying the initial surface areas of primary minerals (which are then changed by the code in proportion to their remaining masses) and describing their dissolution kinetics by means of the TST-based rate law (see Table 7.11). Again, the proton- and water-promoted mechanisms were considered but the hydroxyl-promoted dissolution mechanism was neglected, owing to the

Oxide	Wt%		Element	Mol%
SiO <sub>2</sub>	48.35		Si	17.19
TiO <sub>2</sub>	1.57		Ti	0.42
Al <sub>2</sub> O <sub>3</sub>	15.49		Al	6.49
Fe <sub>2</sub> O <sub>3</sub>	3.26		Fe	3.27
FeO	8.05			
MnO	0.17		Mn	0.05
MgO	7.03		Mg	3.72
CaO	9.92		Ca	3.78
Na <sub>2</sub> O	2.76		Na	1.90
K <sub>2</sub> O	0.51		K	0.23
$P_2O_5$	0.24		Р	0.07
CO,	0.05		С	0.02
H <sub>2</sub> Õ	1.52		0	59.29
-			Н	3.55
Minerals	Mol %	Vol %	Initial moles	Initial surfaces
Plagioclase	59.68	73.86	17.14	3,028.3
Clinopyroxene	21.54	18.32	6.19	751.3
Orthopyroxene	16.02	6.30	4.60	258.2
Magnetite	2.77	1.52	0.80	62.2

TABLE 7.10

Whole-rock chemical analysis of a typical continental tholeiitic flood basalt from the Columbia River Province (Wilson, 1989) and its computed mineralogic composition

## **TABLE 7.11**

endit.

EQ6 input file for modeling high-pressure ( $P_{tot} = P_{CO2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer hosted in continental tholeiitic flood basalts at 60°C in kinetic mode (time frame). (Required input formats are not fully respected.)

The option switch IOPT1 is set to 1 to direct the code to compute the simulation in time frame.

The option switch IOPT4 is set to 1 to activate solid solutions.

The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

enderer										
nmodl1 =	1			nmodl2 = 0	)					
tempc0 = 60.0000E + 00 $tk1 = 0.00000E + 00$				jtemp = 0						
				tk2 = 0.00000E + 00 $tk3 = 0.00000E + 00$						
zistrt=	0.00000H	E+00		zimax= 1	.00000E-	-06				
tstrt=	0.00000I	3+00		timemx = 0	.00000E-	⊦00				
kstpmx≈	500			cplim= (	.00000E-	-00				
dzprnt=	1.00000E	$\pm +00$		dzprlg = 0	.00000E-	+00	ksppmx=	100		
dzplot=	0.10000E	E+00		dzpllg = 0	.00000E-	+00	ksplmx =	10000		
ifile=	60				_		_			
*	1	2	3	4	5	6	7	8	9	10
ioptI-10 =	1	0	0	1	0	0	0	0	0	0
10pt11-20 =	0	0	0	0	0	0	0	0	0	0
10pr1 - 10 =	0	0	0	0	l	0	0	1	0	0
10pr11-20 =	0	0	0	0	0	0	0	0	0	0
10db1-10=	0	0	0	0	0	0	0	0	0	0
*	0	0	0	0	0	0	0	0	0	0
nxopt≖	1									
option=	all									
nxopex=	10									
exception=	Gibbsite									
exception=	Kaolinite	÷								
exception=	Chalcedo	ony								
exception=	Magnetit	e								
exception=	Goethite									
exception=	Dawsoni	te								
exception=	Calcite									
exception=	Magnesi	te								
exception≈	Siderite									
exception=	Dolomite	2								
*	1				•					
species=	$CO_{2}(g)$									
moffg≃	1.00000E	E+01		xlkffg= -	+1.83187	E+00				
*				_						
nrct≈	4									
reactant=	Plagiocla	ise								
jcode=	1			jreac= 0	•					
morr≈	17.1407 <b>E</b>	2+00		modr = 0	.00000E-	-00				
Albite	high	4.53500E-(	)1							
Anorth	ite	5.46500E-(	)1							
endit.										

#### TABLE 7.11 (Continued)

TADLE 7.11 (Commucu)		
nsk = 1	sk = 3028.29E + 00	fk = 0.00000E + 00
nrk = 2	nrpk = -1	
imech=2		
rk0 = 8.03000E - 12	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H^+$	cdac = 0.62600E + 00	
rk0 = 8.35000E - 15	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H_2O$	cdac = 1.00000E + 00	
reactant = Clinopyroxene		
jcode = 1	jreac= 0	
morr = 6.1863E + 00	modr = 0.00000E + 00	
Diopside 7.50000E–01		
Hedenbergite 2.50000E-01		
endit.		
nsk = 1	sk = 751.265E + 00	fk = 0.00000E + 00
nrk = 2	nrpk = -1	
imech= 2		
rk0 = 4.13000E - 10	trk0 = 6.00000E + 01	iact = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H^+$	cdac = 0.70000E + 00	
rk0 = 2.92000E - 15	trk0 = 6.00000E + 01	act = 0
eact = 0.00000E + 00	hact = 0.00000E + 00	
ndact = 1	csigma = 1.00000E + 00	
$udac = H_2O$	cdac = 1.00000E + 00	
reactant = Orthopyroxene		
jcode = 1	Jreac = 0	
morr = 4.60010E + 00	modr = 0.00000E + 00	
Enstatite 6.00000E-01		
Ferrosilite 4.00000E-01		
enuit.	$ab = 258.226E \pm 0.0$	$f_{\rm L} = 0.0000 F \pm 00$
nsk = 1	$s_{\rm K} = 238.2200 \pm 000$	IK- 0.00000E 100
$m_{\rm K} = 2$	mpk- 1	
$r_{\rm rec} = 457000E - 12$	$t_{\rm r} = 6.00000 \pm 0.01$	jact= 0
100 - 4.57000E - 12	hact = 0.00000E + 01	lact = 0
$e_{act} = 0.00000 \pm 00$	csigma = 1.00000E + 00	
udac = H +	cdac = 0.65000E + 00	
$r_{\rm r} = 602000 = 15$	$trk0 = 6.00000E \pm 01$	iact = 0
$IKU = 0.02000E \pm 0.0$	$h_{act} = 0.00000E \pm 01$	
$e_{act} = 0.00000 \pm 00$	aci = 0.00000E + 00	
Huact = 1	$c_{dac} = 1.00000E+00$	
$uual = \Pi_2 O$	Cuac - 1.000001 -00	

TABLE	7.11 (	Continued	)
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Si

electr

6.159324420874708E-04 1.323234709603221E-11

reacta	nt= Magnetite			
jcoo	de = 0	jreac= 0		
mo	rr = 0.79530E + 00	modr = 0.000	)00E+00	
ns	sk = 1	sk= 62.21	180E + 00 fk =	0.00000E+00
n	rk=2	nrpk = -1		
imeo	ch = 2			
rl	x0 = 5.65000E - 13	trk0 = 6.000	000E+01 iact=	0
ea	ct = 0.00000E + 00	hact = 0.000	)00E+00	
nda	ct = 1	csigma = 1.000	)00E+00	
uda	ac = H +	cdac = 0.279	900E+00	
rl	x0 = 3.65000E - 15	trk0 = 5.000	000E+01 iact=	0
ea	ct = 0.00000E + 00	hact = 0.000	000E+00	
nda	ct = 1	csigma = 1.000	)00E+00	
uda *	$ac = H_2O$	cdac = 1.000	000E+00	
dlzic	ip = 0.00000E + 00			
tol	bt = 0.00000E + 00	toldl = 0.000	000E + 00 tolx =	0.00000E+00
tols	at = 0.00000E + 00	tolsst = 0.000	000E + 00	
screw	$v_1 = 0.00000E + 00$	screw2 = 0.000	000E+00 screw3=	0.00000E+00
screw	4 = 0.00000E + 00	screw5 = 0.000	000E+00 screw6=	0.00000E+00
zklog	gu = 0.000	zklogl = 0.000	) zkfac=	0.000
dlzm	c1 = 0.00000E + 00	dlzmx2 = 0.000	000E+00 nordlm=	0
iterm	$\mathbf{x} = 0$	ntrymx = 0		
npslm	$\mathbf{x} = 0$	nsslmx = 0	ioscan=	0
*	·			
* pickup	file written by EQ3NR, v	ersion 7.2b (R139)		
* suppor	ted by EQLIB, version 7.2	2b (R168)		
Water in	equilibrium with analcime	e, calcite, kaolinite,		
chalcedo	ny, daphnite, clinochlore,	magnetite		
at 60øC,	$0.1 \text{ bar PCO}_2$			
endit.				
temp	ci = 6.00000E + 01			
nxmo	d = 0			
iopg	$g_1 = 0$	iopg2=0	iopg3=	0
iopg	g4 = 0	iopg5 = 0	iopg6=	0
iopg	g7 = 0	iopg8 = 0	iopg9=	0
iopgl	10 = 0			
k	ct = 10	ksq = 11	kmt=	11
k	xt = 11	kdim= 11	kprs=	0
0	5.568190070346440E	+01		
AI	2.397285511182285E	-07		
Ca	4.360523578853289E	05		
Cl	3.00000000355616E	-02		
Fe	3.862527706607725E	-06		
Н	1.110727982477859E	+02		
С	5.793621436309842E	-02		
Mg	1.602243822990042E	04		
Na	8.637858497535496E	02		

TABLE /.	11 (Continuea)		
H <sub>2</sub> O	H <sub>2</sub> O	1.744358983526984E+00	
Al <sup>+++</sup>	Al <sup>+++</sup>	-1.710395611381481E+01	
Ca <sup>++</sup>	Ca <sup>++</sup>	-4.562090256614116E+00	
Cl-	$Cl^{-}$	-1.527663478385729E+00	
Fe <sup>++</sup>	Fe <sup>++</sup>	-6.567807046642986E+00	
$H^+$	$H^+$	-7.592429061239226E+00	
HCO <sub>3</sub>	$HCO_3^-$	-1.269686271112627E+00	
Mg <sup>++</sup>	Mg <sup>++</sup>	-3.959985610775256E+00	
Na <sup>+</sup>	Na <sup>+</sup>	-1.076209250999970E+00	
SiO <sub>2</sub> (aq)	SiO <sub>2</sub> (aq)	-3.23070000000000E+00	
O <sub>2</sub> (g)	O <sub>2</sub> (g)	-6.098389925982678E+01	



Figure 7.13. Moles of reactant minerals destroyed during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts as a function of time and the reaction progress variable.

low pH values constrained by  $CO_2$  injection (see below). As in previous cases, selected secondary minerals were assumed to attain instantaneous partial equilibrium, owing to the difficulty in determining the surface area for solid phases which are not present in the considered system at the beginning of the simulation.

The diagrams of Figures 7.13 to 7.18 depict the main results of this reaction path modelling exercise against both the reaction progress variable,  $\xi$ , and time.



Figure 7.14. Amounts of product minerals formed during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts as a function of time and the reaction progress variable. Letters D, S, C and Dw mark the onset of the precipitation of secondary dolomite, siderite, calcite and dawsonite, respectively.

#### 7.4.2. Solid reactants

Throughout CO<sub>2</sub> injection coupled with basalt dissolution, the aqueous solution remains strongly undersaturated with respect to plagioclase ( $A \le -9.6$  kcal mol<sup>-1</sup>), clinopyroxene ( $A \le -16.3$  kcal mol<sup>-1</sup>), orthopyroxene ( $A \le -8.3$  kcal mol<sup>-1</sup>) and magnetite ( $A \le -3.4$  kcal mol<sup>-1</sup>). Owing to these relatively low values of A, the dissolution rates of primary phases are always independent of thermodynamic affinity (*see* Section 6.4.2) and depend on pH only.

The moles of primary minerals consumed during continuous  $CO_2$  injection and basalt dissolution are presented in Figure 7.13. The two most abundant solid phases, plagioclase and clinopyroxene, contribute most chemical components to the considered system, but their contributions are generally reversed with respect to their surface areas, owing to the overwhelming weight of the dissolution rate (in mol cm<sup>-2</sup> s<sup>-1</sup>), which is much higher for clinopyroxene than for the labradoritic plagioclase, in acidic solutions (see Tables 6.1 and 7.11). Only in the final part of the simulation, namely upon exhaustion of clinopyroxene, plagioclase becomes the top provider of chemical components.



Figure 7.15. Concentrations of solutes during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts as a function of time and the reaction progress variable. Letters D, S, C and Dw mark the onset of the precipitation of secondary dolomite, siderite, calcite and dawsonite, respectively, as shown in Figure 7.14.

The contributions of orthopyroxene and magnetite are generally subordinate owing to their comparatively low dissolution rates (in mol  $cm^{-2} s^{-1}$ ) and amounts; however, orthopyroxene becomes an important supplier of chemical components in the final part of the simulation, upon exhaustion of other solid reactants.

Clinopyroxene is totally consumed after 26,100 years, for  $\xi$  of 22.9 mol, plagioclase is exhausted after 191,300 years, for  $\xi$  of 28.4 mol, and magnetite is completely consumed after 280,600 years, for  $\xi$  of 28.6 mol. The simulation ends for  $\xi$  of 28.7 mol, upon exhaustion of orthopyroxene.

## 7.4.3. Solid product phases

The solid phases which are precipitated during continuous  $CO_2$  injection in a deep aquifer hosted in tholeiitic continental flood basalts are, in order of appearance, chalcedony, kaolinite, goethite, dolomite, siderite, calcite and dawsonite [NaAlCO<sub>3</sub>(OH)<sub>2</sub>] (Figure 7.14). Both chalcedony and kaolinite form immediately upon  $CO_2$  injection, goethite production begins after 9 h and 10 min ( $\xi = 5.76 \mu$ mol), whereas the precipitation of carbonate minerals starts much later: that of dolomite after 1 year ( $\xi = 5.12 \text{ mmol}$ ),



Figure 7.16. Changes in (a) pH and (b)  $f_{O_2}$  during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts as a function of time and the reaction progress variable. Letters G, D, S, C and Dw mark the onset of the precipitation of secondary goethite, dolomite, siderite, calcite and dawsonite, respectively.



Figure 7.17. High-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts: masses of CO<sub>2</sub> sequestered through dissolution in the aqueous solution (solubility trapping) and incorporation in solid carbonates (mineral fixation), and total mass of sequestered CO<sub>2</sub> as a function of time and the reaction progress variable. Letters D, S, C and Dw mark the onset of the precipitation of secondary dolomite, siderite, calcite and dawsonite, respectively, as shown in Figure 7.14.

that of siderite after 1 year and 3 months ( $\xi = 5.12$  mmol), that of calcite after 8 years and 5 months ( $\xi = 39.8$  mmol) and that of dawsonite after 53 years and 5 months ( $\xi = 227$  mmol).

The log-moles of each solid phase increase close to linearly with log  $\xi$  after an initial non-linear trend, which is evident for carbonate minerals only in Figure 7.14. Chalcedony is by far the most important solid product phase throughout the process, carbonate minerals and kaolinite are produced in similar amounts (after an initial settling-down period), whereas goethite is the least important secondary phase.

The masses of produced carbonate minerals amount to: (i) 91.7 g of dolomite, 31.9 g of calcite, 20.9 g of siderite and 16.6 g of dawsonite, for a total of 161.1 g, after 255 years ( $\xi = 1 \text{ mol}$ ); (ii) 736 g of dolomite, 355 g of calcite, 180 g of siderite and 288 g of dawsonite, for a total of 1,559 g, after 4,080 years ( $\xi = 10 \text{ mol}$ ); (iii) 1,015 g of dolomite, 871 g of calcite, 319 g of siderite and 960 g of dawsonite, for a total of 3,165 g, after 26,100 years ( $\xi = 22.9 \text{ mol}$ ), upon exhaustion of clinopyroxene. These figures confirm the high potential of CO<sub>2</sub> sequestration of these basaltic rocks, including the important role of dawsonite, in spite of the long time required for mineral fixation.



Figure 7.18. Variation of the  $\Delta V/V$  parameter as a function of both reaction progress and time during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a deep aquifer ( $T = 60^{\circ}$ C) hosted in continental tholeiitic flood basalts. Letters D, S, C and Dw mark the onset of the precipitation of secondary dolomite, siderite, calcite and dawsonite, respectively, as shown in Figure 7.14.

## 7.4.4. The aqueous solution

Also throughout this computer experiment, the main solute is aqueous  $CO_2$ , with a concentration varying between 48,100 and 47,200 ppm (Figure 7.15). In other words,  $CO_2$  concentration remains practically constant at 1.1 mol kg<sup>-1</sup>, the same figure of the serpentinite experiment (*see* Section 7.3.2.4.), owing to the constancy in  $f_{CO_2}$ , at fixed temperature and pressure.

Aqueous CO<sub>2</sub> is followed in order of decreasing importance (at least for  $\xi > 0.01$  mol) by HCO<sub>3</sub><sup>-</sup>, Na<sup>+</sup>, Cl<sup>-</sup>, Ca<sup>2+</sup>, Fe<sup>2+</sup>, SiO<sub>2</sub> and Mg<sup>2+</sup>. Dissolved SiO<sub>2</sub> is kept constant throughout the simulation by saturation with chalcedony, a condition which is attained immediately upon CO<sub>2</sub> injection. Chloride concentration remains constant at the initial value, 1,060 mg kg<sup>-1</sup>, for over 22 years and then it increases. Such an increase is initially weak but becomes more and more important with time, especially after 4,000 years ( $\xi > 10$  mol). This effect is caused by the progressive incorporation of water in precipitating hydrous minerals (kaolinite, dawsonite and goethite), a phenomenon known as dry-up (Reed, 1997), as already recalled above. The concentrations of all the other solutes, i.e.  $HCO_3^-$ ,  $Na^+$ ,  $Ca^{2+}$ ,  $Fe^{2+}$  and  $Mg^{2+}$ experience an initial increase (very pronounced for  $Ca^{2+}$ ,  $Fe^{2+}$  and  $Mg^{2+}$  and rather weak for  $HCO_3^-$  and  $Na^+$ ), before the attainment of saturation with respect to carbonate minerals. This is followed by a general stabilization after the onset of dawsonite precipitation (i.e. after 53 years and 5 months, for  $\xi$  of 227 mmol, point Dw in Fig. 7.15), with more or less important changes in the intermediate period. It must be underscored that for  $\xi > 227$ mmol, calcite, dolomite, siderite and dawsonite are in equilibrium with the aqueous solution, a condition which fixes the activities of free  $Ca^{2+}$ ,  $Mg^{2+}$ ,  $Fe^{2+}$  and  $Na^+$ , respectively. Only limited changes in the total concentrations of these solutes are possible, owing to variations in salinity and consequent changes in the concentrations of complex species.

The simulation predicts that upon  $CO_2$  injection, the pH of the aqueous solution attains immediately the value of 4.87 (Fig. 7.16a), which is higher by one unit than the initial pH value of the serpentinite case, 3.83, and comparable with its final pH value, 4.85 (*see* Section 7.3.2.4). This relatively high initial pH value of the basalt case is consistent with the high concentrations of Na<sup>+</sup> and HCO<sub>3</sub><sup>-</sup> in the aqueous solution, which are the two prevailing solutes (see above). During progressive  $CO_2$  injection and basalt dissolution, the pH of the aqueous solution increases up to the maximum value of 5.12 at point Dw and remains stable afterwards, since it is buffered by coexistence of carbonate minerals, apart from a minor decrease to 5.07 in the final part of the simulation.

Changes in log  $f_{O_2}$  are similar to those of the serpentinite case, with a sharp increase from -59.8 to -37.6 concurrent with the attainment of goethite saturation (Fig. 7.16b, point G), which is followed by a gradual decrease until the aqueous solution saturates with siderite (point S). From this point, coexistence of goethite and siderite at constant  $f_{CO_2}$ , temperature and pressure (reaction 7-47) fixes the log  $f_{O_2}$  at -47.53, as already mentioned in Section 7.3.2.4.

## 7.4.5. The CO<sub>2</sub> sequestration

The masses of  $CO_2$  sequestered through both solubility trapping (i.e. as dissolved  $CO_2$  and related carbonate aqueous species) and mineral fixation (i.e. the  $CO_2$  stored in secondary carbonate minerals, namely dolomite, siderite, calcite and dawsonite in the considered case), as well as their sum are shown in Figure 7.17 as a function of the reaction progress and time.

As already discussed above,  $CO_2$  sequestration through solubility trapping is instantaneous, but quite small, as it amounts to ~50 g in 1 kg of water at the specified conditions of 60°C and  $P_{tot} = P_{CO_2} = 100$  bar. Incorporation of  $CO_2$  in product carbonates begins at the onset of dolomite precipitation (point D) and continues with the contributions of siderite from point S, calcite from point C and dawsonite from point Dw. Although mineral fixation is a late process, it becomes more and more important with time and the overall sequestration potential of the process is considerable, with 1,580 g kg<sup>-1</sup> of water upon exhaustion of primary minerals. However, the contribution of mineral fixation equals that of solubility trapping only after 175 years, against the 5 years of the serpentinite case (*see* Section 7.3.2.5).

## 7.4.6. Changes in the porosity of aquifer rocks

Also for the tholeiitic basalt case, the  $\Delta V/V$  parameter was computed and plotted as a function of both the reaction progress and time (Fig. 7.18) to estimate the variations in the porosity of aquifer rocks consequent to high-pressure CO<sub>2</sub> injection (*see* equation (7-48) and Section 7.3.2.6). The  $\Delta V/V$  parameter exhibits a sharp increase when carbonate minerals begin to precipitate, achieves the 30% value soon after point C, where the aqueous solution saturates with calcite after 8 years and 5 months ( $\xi = 39.8$  mmol), and attains the maximum value close to 50% after 250–1,500 years. Considering that the 30% value represents a sort of upper threshold even for high-porosity rocks, continuous CO<sub>2</sub> injection in a deep aquifer hosted in tholeiitic continental flood basalts is expected to reduce dramatically its porosity. The long-term sustainability of this process is therefore dubious.

These volume changes are not surprising, since the considered primary phases (plagioclase, clinopyroxene, orthopyroxene and magnetite) have very low molar volumes, a situation quite similar to the dunite example of Xu et al. (2000, 2004).

All in all, the basalt example provide contrasting results, as the positive findings concerning the potential of  $CO_2$  sequestration of Na-bearing minerals and rocks, in the form of secondary dawsonite, are in striking contrast with the negative findings regarding the huge reduction in the effective porosity of the aquifer.

## 7.5. Reaction path modelling of geological CO<sub>2</sub> sequestration in basaltic glass

In Section 7.4, the continental tholeiitic flood basalt involved in  $CO_2$  sequestration was considered holocrystalline, that is composed entirely of crystals. This is a limiting, somewhat unlikely case for volcanic rocks, which are usually composed by varying amounts of glass, mostly depending on their cooling history and age. Glasses are typically formed by quenching of silicate melts that prevents the organization of the random liquid structure into ordered crystalline structures. Glasses are metastable and are doomed to crystallize, soon or later. They are often believed to be more reactive than their crystalline counterparts, but this is not always the case, as discussed in Section 6.6.6.

The dissolution kinetics of silicate glasses has received considerable attention (*see* Sections 6.3.2 and 6.6.6) and, in particular, that of a basaltic glass from Stapafell mountain in southwestern Iceland was investigated experimentally and described in the framework of the multi-oxide dissolution model by Oelkers, Gislason and coworkers (Section 6.6.6.2). As already recalled, the chemical composition normalized to 1 Si atom of this basaltic glass is Si  $Ti_{0.02} Al_{0.36} Fe(III)_{0.02} Fe(II)_{0.17} Mg_{0.28} Ca_{0.26}Na_{0.08} K_{0.008} O_{3.38}$ . As underscored by Oelkers and Gislason (2001), the basaltic glass from Stapafell is similar in chemical composition to both the average oceanic crust [Si  $Ti_{0.02} Al_{0.37} Fe(III)_{0.04} Fe(II)_{0.13} Mg_{0.23} Ca_{0.27} Na_{0.10} K_{0.008} O_{3.35}$ ] and the average mid ocean ridge basalt [Si  $Ti_{0.02} Al_{0.36} Fe(III)_{0.14} Mg_{0.28} Ca_{0.25} Na_{0.10} K_{0.003} O_{3.33}$ ], which is usually indicated by the acronym MORB. Note that since the separate contents of Fe(III) and Fe(II) are not reported for the average MORB, its Fe(III)/Fe(II) ratio was assumed to be equal to that of

average oceanic crust to distribute total Fe between the ferric and ferrous reservoirs. The basaltic glass from Stapafell mountain is also quite similar to the continental tholeiitic flood basalt considered in Section 7.4, whose chemical composition normalized to 1 Si atom is Si  $Ti_{0.02} Al_{0.38}$  Fe(III)<sub>0.05</sub> Fe(II)<sub>0.14</sub> Mg<sub>0.22</sub>Ca<sub>0.22</sub> Na<sub>0.11</sub> K<sub>0.01</sub> O<sub>3.34</sub>.

In spite of the compositional similarity, the basaltic glass from Stapafell mountain is obviously different from the holocrystalline continental tholeiitic flood basalt considered in previous section from the mineralogical point of view and, consequently, a different reaction path is expected upon interaction with water and  $CO_2$ . To evaluate these differences, in this section we simulate the geological storage of  $CO_2$  in a hypothetical aquifer hosted in the basaltic glass from Stapafell mountain.

#### 7.5.1. Setting up the water–rock interaction model

## 7.5.1.1. The need for two solid reactants

As anticipated in Section 6.6.6.2, Gislason and Oelkers (2003) considered the hydrolysis reaction (6-142) for the hydrated basaltic glass of composition  $SiAl_{0.36}O_2(OH)_{1.08}$ , which presumably forms a thin layer at the surface of the basaltic glass during its dissolution. Incidentally, note that the chemical formula of hydrated basaltic glass can also be expressed as  $SiO_2 \cdot 0.18Al_2O_3 \cdot 0.54 H_2O$ , in terms of the component oxides, or  $SiO_2 \cdot 0.36Al(OH)_3$ , in terms of the hypothetical component minerals. The equilibrium constants of the hydrolysis reaction of hydrated basaltic glass (reaction 6-142) was evaluated by Gislason and Oelkers (2003) based on the latter chemical formula. First, they assumed that amorphous silica and gibbsite are the two hypothetical component minerals of hydrated basaltic glass, in a molar ratio 1:0.36. Then, they computed the equilibrium constants of the hydrolysis reaction of hydrated basaltic glass by summing log  $K_{sp, amorphous}$  silica plus 0.36 log  $K_{sp, gibbsite}$ . The obtained log  $K_{sp}$  values of hydrated basaltic glass at varying temperatures are reported in Section 6.6.6.2.

Before running EQ6, the log  $K_{sp}$  of hydrated basaltic glass were re-computed at the temperature, pressure grid required by EQ3/6, through simple interpolation, and inserted in the thermodynamic database of EQ3/6. After this necessary preparatory step, the saturation state of any aqueous solution with respect to this hydrated basaltic glass can be computed by means of EQ3/6. Moreover, this hydrated basaltic glass can be treated as a pure mineral reactant and its dissolution can be properly simulated by running EQ6 (Table 7.12).

To this purpose, the dissolution rate law proposed by Gislason and Oelkers (2003), equation (6-143), is inserted in the EQ6 input file as a TST rate law involving the activities of both H<sup>+</sup> and Al<sup>3+</sup> ions, as we have already seen in the albite example (*see* Section 7.2.4). It must be recalled that Gislason and Oelkers (2003) report the dissolution rate of the basaltic glass normalized to its geometric surface area,  $r_{+,GEOM}$ . However, they also underscore that these  $r_{+,GEOM}$  are 92 times larger than the dissolution rates normalized to the BET surface area,  $r_{+,BET}$ . Both values were inserted in two separated EQ6 input files, to evaluate the effects of these limiting dissolution rates on the EQ6 simulation.

Only Al and Si are released to the aqueous solution through dissolution of the hydrated basaltic glass involved so far in the model. In contrast, we know that dissolution

## **TABLE 7.12**

EQ6 input file for modeling high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a hypothetical deep aquifer hosted in basaltic glass at 60°C in kinetic mode (time frame). The rk0 value of  $2.52370 \times 10^{-10}$  mol cm<sup>-2</sup> s<sup>-1</sup> is the dissolution rate constant normalized to the geometric surface of the basaltic glass at 60°C (from Oelkers and Gislason, 2003). (Required input formats are not fully respected.)

The option switch IOPT1 is set to 1 to direct the code to compute the simulation in time frame. The option switch IOPT4 is set to 1 to activate solid solutions.

The print option switch IOPR8 is set to 1 to direct the code to print a table of equilibrium gas fugacities at each print point.

endit.

nmodl1 =	1			nmodl2 =	0					
tempc0=	60.0000E	E+00		jtemp=	0					
tk1=	0.00000E	E+00		tk2=	0.00000E	+00	tk3=	0.00000E	+00	
zistrt=	0.00000E	E + 00		zimax=	1.00000E	-01				
tstrt=	0.00000E	E + 00		timemx=	0.00000E	+00				
kstpmx=	500			cplim=	0.00000E	+00				
dzprnt=	1.00000E	E-02		dzprlg=	0.00000E	+00	ksppmx=	100		
dzplot=	0.10000E	E + 00		dzpllg=	0.00000E	+00	ksplmx =	10000		
ifile=	60									
*	1	2	3	4	5	6	7	8	9	10
iopt1-10=	1	0	0	1	0	0	0	0	0	0
iopt11-20=	0	0	0	0	0	0	0	0	0	0
iopr1-10=	0	0	0	0	1	0	0	1	0	0
iopr11-20=	0	0	0	0	0	0	0	0	0	0
iodb1-10=	0	0	0	0	0	0	0	0	0	0
iodb11-20=	0	0	0	0	0	0	0	0	0	0

nxopt= 1

option = all

nxopex = 10

exception= Gibbsite

exception= Kaolinite

exception= Chalcedony

exception= Dawsonite exception= Calcite exception= Magnesite

exception = Siderite

exception= Magnetite exception= Goethite

nng =	1
species =	$CO_2(g)$
moffg=	1.00000E+01

xlkffg= +1.83187E+00

nrct= 2 reactant= Hyd-basalt-glass

jcode= 0 morr= 56.09000E+00 nsk= 1

jreac= 0 modr= 0.00000E+00 sk= 4100.00E+00

fk = 0.00000E + 00

```
TABLE 7.12 (Continued)
```

	(communed)					
nrk=	2	nrpk=	-1			
imech=	1	•				
rk0=	2.52370E-10	trk0=	6.00000E+01	iact=	0	
eact=	0.00000E + 00	hact=	0.00000E+00			
ndact=	2	csigma=	1.00000E+00			
udac=	$H^+$	cdac=	1.00000E+00			
udac=	Al+++	cdac=	-0.33333E+00			
*						
reactant=	rock					
jcode=	2	jreac=	0			
morr=	56.09000E+00	modr=	0.00000E + 00			
vreac=	0.00000E + 00					
Fe	1.9000E-01					
Mg	2.8000E-01					
Ca	2.6000E-01					
Na	8.0000E-02					
0	7.8000E-01					
endit.						
nsk=	0	sk=	0.00000E+00	fk=	0.00000E + 00	
nrk=	1	nrpk=	0			
rk1=	1.00000E+00	rk2=	0.00000E+00	rk3=	0.00000E+00	
*						
dlzidp=	0.00000E + 00					
tolbt=	0.00000E + 00	toldl=	0.00000E + 00	tolx=	0.00000E + 00	
tolsat=	$0.00000E \pm 00$	tolsst=	0.00000E + 00			
screw1=	$0.00000E \pm 00$	screw2=	0.00000E + 00	screw3=	0.00000E + 00	
screw4=	0.00000E + 00	screw5=	0.00000E + 00	screw6=	0.00000E + 00	
zklogu=	0.000	zklogl=	0.000	zkfac =	0.000	
dlzmx1 =	0.00000E + 00	dlzmx2=	0.00000E + 00	nordlm=	0	
itermx =	0	ntrymx=	0			
npslmx=	0	nsslmx=	0	ioscan=	0	
* pickup file * supported b Water in equi chalcedony, d	written by EQ3NR, y EQLIB, version 7 librium with analcin aphnite, clinochlore	version 7.2b (R13 7.2b (R168) me, calcite, kaolini e, magnetite	9) te,			
11 0000C, 0.1	oar PCO <sub>2</sub>					
tomnoi —	6 00000E ± 01					
tempci=	0.00000E+01					
nxmoa=	0		0		0	
				·	<i>·</i> · ·	

	iopg1 = 0 iopg4 = 0 iopg7 = 0	iopg2=0 iopg5=0 iopg8=0	iopg3 = 0 iopg6 = 0 iopg0 = 0
	$\log 7 = 0$	lopga- 0	lopg9= 0
	kct = 10	ksq = 11	kmt = 11
	kxt = 11	kdim = 11	kprs = 0
0	5.568190070346440E+01		-
Al	2.397285511182285E-07		
Ca	4.360523578853289E-05		
Cl	3.00000000355616E-02		
Fe	3.862527706607725E-06		

Н	1.110727982477859E+02	
С	5.793621436309842E-02	
Mg	1.602243822990042E04	
Na	8.637858497535496E-02	
Si	6.159324420874708E-04	
electr	1.323234709603221E-11	
H <sub>2</sub> O	H <sub>2</sub> O	$1.744358983526984E \pm 00$
Al <sup>+++</sup>	Al <sup>+++</sup>	- 1.710395611381481E+01
Ca <sup>++</sup>	Ca <sup>++</sup>	-4.562090256614116E+00
C1-	Cl <sup>-</sup>	- 1.527663478385729E+00
Fe <sup>++</sup>	Fe <sup>++</sup>	$-6.567807046642986E \pm 00$
$H^+$	H <sup>+</sup>	-7.592429061239226E+00
HCO <sub>3</sub>	HCO <sub>3</sub>	- 1.269686271112627E+00
Mg <sup>++'</sup>	Mg <sup>++</sup>	-3.959985610775256E+00
Na <sup>+</sup>	Na <sup>+</sup>	-1.076209250999970E+00
SiO <sub>2</sub> (aq)	SiO <sub>2</sub> (aq)	-3.23070000000000E+00
O <sub>2</sub> (g)	O <sub>2</sub> (g)	- 6.098389925982678E+01

TABLE 7.12 (Continued)

of basaltic glass adds also several other metals to the aqueous phase. To take into account the fate of all these metals, it is necessary to involve a second reactant in the EQ6 simulation. This second reactant can be treated as a so-called *special reactant* (in EQ6 meaning), that is a substance whose chemical composition is known but whose thermodynamic properties are not. Obviously, substances of this kind (e.g., any rock) cannot be inserted in the thermodynamic database of EQ3/6.

In the basaltic glass case, the special reactant has composition  $Ti_{0.02}$  Fe(III)<sub>0.02</sub> Fe(III)<sub>0.17</sub> Mg<sub>0.28</sub> Ca<sub>0.26</sub> Na<sub>0.08</sub> K<sub>0.008</sub> O<sub>0.82</sub>, since it is the Si-free, Al-free portion of the whole basaltic glass. Titanium and K are not considered in the EQ6 simulation owing to the low solubility of Ti minerals and the low content of K, which was also neglected in the tholeiitic basalt case. The number of oxygen atoms have to be decreased, therefore, to 0.78 to balance the metal charges. Note that it is very important to maintain the chemical formula of special reactants stoichiometrically balanced by adjusting the number of oxygen atoms, otherwise incorrect  $f_{O_2}$  values are computed by EQ6. Besides, it is necessary to instruct EQ6 to dissolve this special reactant at the same rate of the hydrated basaltic glass. This is accomplished by setting the rsk1 flag to 1.

## 7.5.1.2. The molar volume of basaltic glass

Let us assume again that the considered system comprises 1,000 g of water and has an initial porosity of 0.3. Equation (7-44) dictates that initially basaltic glass occupies a volume of 2,333 cm<sup>3</sup>. The molar volume of the basaltic glass is needed to compute both the moles of basaltic glass initially present in the system and the changes in the  $\Delta V/V$ parameter during the simulation by using equation (7-48).

The molar volume of the basaltic glass can be calculated based on the partial molar volumes and the volumetric thermal expansion coefficients of the component oxides, whereas their compressibilities can be neglected at the relatively low pressures of interest.

These volumetric parameters of the component oxides of silicate melts and glasses have been the subject of several investigations (e.g., Bottinga et al., 1982; Lange and Carmichael, 1987, 1990; Knoche et al., 1995; Lange, 1997; Toplis and Richet, 2000). Inserting the partial molar volumes  $\overline{V_i}$  (cm<sup>3</sup> mol<sup>-1</sup>) and the thermal expansions  $d\overline{V_i}/dT$ (cm<sup>3</sup> mol<sup>-1</sup> K<sup>-1</sup>) given by Lange and Carmichael (1987) into the simplified relation (*T* in *K*):

$$v_{\text{liq}} = \sum_{i} X_{i} \cdot \left[ \overline{V}_{i,1673K} + \frac{d\overline{V}_{i}}{dT} \cdot (T - 1673) \right]$$
(7-49)

the molar volume of the basaltic glass of interest at 60°C, 1 bar results to be 21.05 cm<sup>3</sup> mol<sup>-1</sup>, corresponding to 41.60 cm<sup>3</sup> mol<sup>-1</sup> for the chemical formula normalized to 1 Si atom. Therefore, the moles of basaltic glass initially contained in the considered system amount to 2,333 cm<sup>3</sup>/41.60 cm<sup>3</sup> mol<sup>-1</sup> = 56.09 mol. Some words of caution are needed on the possible contribution of water to the molar volume of the basaltic glass. Water was not considered in previous calculations since its content in the basaltic glass of Stapafell is not reported by Oelkers and Gislason (2001) and Gislason and Oelkers (2003). Owing to its low molecular weight (18.015 g mol<sup>-1</sup>) compared to other component oxides of silicate melts and glasses, relatively small amounts by weight of water in silicate glasses is relatively low (Richet and Polian, 1998 estimated a value of  $12 \pm 0.5$  cm<sup>3</sup> mol<sup>-1</sup> for andesite glasses), the molar volume estimated for the anhydrous basaltic glass of Stapafell is doomed to decrease, albeit to an unknown extent, upon consideration of water.

#### 7.5.1.3. Completion of the EQ6 input file

The preparation of the input file for the EQ6 simulation is completed by setting temperature at 60°C,  $f_{CO_2}$  at 67.9 bar (which means to assume injection of pure CO<sub>2</sub> at a constant pressure of 100 bar) and the surface area of hydrated basalt glass in contact with 1,000 g of water at the initial value of 4,100 cm<sup>2</sup> (which is automatically changed by EQ6 proportionally to its remaining mass). Note that the values of these variable are the same as in previous cases. As in the tholeiitic basalt run, the initial aqueous solution was assumed to be in equilibrium with analcime, calcite, kaolinite, daphnite, clinochlore, chalcedony and magnetite at 60°C, 1 bar under a  $f_{CO_2}$  of  $10^{-1}$  bar (see Section 7.1.2).

Again, instantaneous precipitation was hypothesized for the secondary solid phases, as it is difficult to constrain the initial value of the surface area for minerals which are not present in the system at the beginning of the simulation.

# 7.5.2. The time scales

The main results of reaction path modelling are depicted against both the reaction progress variable,  $\xi$ , and time in the diagrams of Figures 7.19 to 7.23, in which two time scales are reported. They refer to the dissolution rate of the basaltic glass normalized to either the geometric surface area,  $r_{+,\text{GEOM}}$ , or the BET surface area,  $r_{+,\text{BET}}$ . As expected on



Figure 7.19. Amounts of secondary product phases generated during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a hypothetical deep aquifer ( $T = 60^{\circ}$ C) hosted in basaltic glass as a function of time and the reaction progress variable. Letters S, D, Dw and M mark the onset of the precipitation of secondary siderite, dolomite, dawsonite and magnesite, respectively.

the basis of equation (7-16), the  $r_{+,\text{GEOM}}/r_{+,\text{BET}}$  ratio of 92 constrains the time<sub>rate,GEOM</sub>/ time<sub>rate,BET</sub> ratio at 1/92 and brings about a corresponding shift in the two time scales.

The results of the two separate simulations (concentrations of solutes, pH of the aqueous phase, oxygen fugacity, moles of secondary solid phases, etc.) are exactly the same for any given value of the reaction progress variable,  $\xi$ . However, the same values of  $\xi$  are attained at different times in the two simulations.

This example demonstrates once more that it is advisable to consider  $\xi$  as the true master variable, owing to possible uncertainties in the time scale brought about by poor knowledge of the dissolution rates and/or reactive surface areas of solid phases.

# 7.5.3. Solid product phases

The secondary carbonate minerals generated during high-pressure  $CO_2$  injection in a hypothetical deep aquifer hosted in basaltic glass are, in order of appearance, siderite, dolomite, dawsonite and magnesite (Fig. 7.19). The precipitation of these solid phases



Figure 7.20. Concentrations of solutes during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a hypothetical deep aquifer ( $T = 60^{\circ}$ C) hosted in basaltic glass as a function of time and the reaction progress variable. Letters S, D, Dw and M mark the onset of the precipitation of secondary siderite, dolomite, dawsonite and magnesite, respectively, as shown in Figure 7.19.

begins at values of  $\xi$  indicated by the labels S, D, Dw and M, respectively. These labels are also reported in the following plots. The most important of these solid phases, in terms of mass, is dolomite followed by siderite, dawsonite and magnesite.

Comparison of Fig. 7.19 with Fig. 7.14 shows that magnesite precipitation occurs in the basaltic glass run instead of calcite production during the tholeiitic basalt example. At first sight, this difference might seem surprising, since the Ca/Mg molar ratio of the basaltic glass, 0.93, is similar to that of the tholeiitic basalt, 1.000. In order to explain this difference in secondary carbonate minerals let us consider the two dissolution processes in detail.

As discussed in Section 7.4.2, the dissolution of the holocrystalline tholeiitic basalt is dominated by destruction of clinopyroxene (10% of Ca and 7.5% of Mg on an elemental basis) and plagioclase (4.2% of Ca and no Mg), with a subordinate contribution of orthopyroxene (12% of Mg and no Ca), apart from the last steps of the simulation, upon exhaustion of clinopyroxene, when orthopyroxene dissolution becomes important. This


Figure 7.21. Changes in (a) pH and (b)  $f_{O_2}$  during high-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a hypothetical deep aquifer ( $T = 60^{\circ}$ C) hosted in basaltic glass as a function of time and the reaction progress variable. Letters G, S, D, Dw and M mark the onset of the precipitation of secondary goethite, siderite, dolomite, dawsonite and magnesite, respectively.



Figure 7.22. High-pressure ( $P_{tot} = P_{CO_2} = 100$  bar) CO<sub>2</sub> injection in a hypothetical deep aquifer ( $T = 60^{\circ}$ C) hosted in basaltic glass: masses of CO<sub>2</sub> sequestered through dissolution in the aqueous solution (solubility trapping) and incorporation in solid carbonates (mineral fixation), and total mass of sequestered CO<sub>2</sub> as a function of time and the reaction progress variable. Letters S, D, Dw and M mark the onset of the precipitation of secondary siderite, dolomite, dawsonite and magnesite, respectively, as shown in Figure 7.19.

larger availability of Ca with respect to Mg during the dissolution of tholeiitic basalt is in line with the production of calcite and dolomite in similar amounts.

In contrast, dissolution of basaltic glass contributes chemical constituents to the aqueous phase in the same stoichiometric proportions of the glass itself, i.e. with a Ca/Mg molar ratio of 0.93. This explains the overwhelming production of dolomite accompanied by subordinate magnesite.

Comparison of Fig. 7.19 with Fig. 7.14 also shows that during the dissolution of the basaltic glass, precipitating carbonate minerals are accompanied by secondary chalcedony, kaolinite and goethite in amounts comparable with those of the tholeiitic basalt case.

# 7.5.4. The aqueous solution

Inspection of Figs. 7.20 and 7.15 indicates that the concentrations of aqueous CO<sub>2</sub>,  $HCO_3^-$ ,  $Na^+$ ,  $C1^-$ ,  $Fe^{2+}$  and  $SiO_2$  during high-pressure CO<sub>2</sub> injection in a hypothetical deep aquifer hosted in basaltic glass are comparable with those of the tholeiitic basalt simulation. Also the concentration of dissolved  $Al^{3+}$  (not shown) is similar in the two runs. This



Figure 7.23. Variation of the  $\Delta V/V$  parameter as a function of both reaction progress and time during highpressure ( $P_{\text{tot}} = P_{\text{CO}_2} = 100 \text{ bar}$ ) CO<sub>2</sub> injection in a hypothetical deep aquifer ( $T = 60^{\circ}$ C) hosted in basaltic glass. Letters S, D, Dw and M mark the onset of the precipitation of secondary siderite, dolomite, dawsonite and magnesite, respectively, as shown in Figure 7.19.

is due to the fact that imposed temperature and  $f_{CO_2}$  and the relevant product solid phases (chalcedony, kaolinite, siderite and dawsonite) are the same.

In contrast, the aqueous solution interacting with the basaltic glass is richer in dissolved  $Mg^{2+}$  and poorer in  $Ca^{2+}$  with respect to the aqueous phase circulating into the tholeiitic basalt. This difference in dissolved  $Mg^{2+}$  and  $Ca^{2+}$ concentrations reflects the different dissolution behaviour of the two basaltic rocks, as discussed in the previous section.

Not surprisingly, also the pH values during the basaltic glass simulation span a limited range, from 4.87 to 5.13, (Fig. 7.21a) which coincides with that of the tholeiitic basalt run, from 4.87 to 5.12 (Fig. 7.16a).

The  $f_{O_2}$  values predicted by the two simulations are also identical (see Figs. 7.21b and 7.16b), and are largely controlled by precipitation of goethite (after point G) and coexistence of goethite and siderite (after point S), as already noted above.

# 7.5.5. The CO<sub>2</sub> sequestration

Figure 7.22 depicts the amounts of  $CO_2$  sequestered through both solubility trapping (i.e. as aqueous  $CO_2$  and other dissolved carbonates) and mineral fixation (i.e. the  $CO_2$  stored in dolomite, siderite, dawsonite and magnesite), as well as the total mass of sequestered  $CO_2$  as a function of the reaction progress. The three  $CO_2-\xi$  curves are almost identical to those of the tholeiitic basalt case (Fig. 7.17) and the same considerations of Section 7.4.5 apply also to the basaltic glass example.

Considering the process with respect to time, it must be underscored that the time scale of basaltic glass carbonation, based on the results obtained for  $r_{+,BET}$ , is comparable to that of the carbonation of holocrystalline tholeiitic basalt. In contrast, basaltic glass carbonation occurs two orders of magnitude faster than the carbonation of holocrystalline tholeiitic basalt if the results computed for  $r_{+,GEOM}$  are taken into account. This matter requires further investigation for its important practical implications, in that basaltic glass and other silicate glasses could be very interesting materials for the geological storage of CO<sub>2</sub> and mineral carbonation, if the  $r_{+,GEOM}$  values were representative of their dissolution rates.

# 7.5.6. Changes in the porosity of aquifer rocks

During high-pressure CO<sub>2</sub> injection in a hypothetical deep aquifer hosted in basaltic glass the  $\Delta V/V$  parameter experiences a pronounced increase upon the onset of siderite and dolomite precipitation (points S and D, respectively) and a corresponding marked decrease in aquifer porosity is expected. Upon the onset of dawsonite precipitation (point Dw), the  $\Delta V/V$  parameter exceeds 30%, which is a sort of upper threshold even for high-porosity rocks. Therefore, total clogging of pore spaces and voids is expected after point Dw.

Although the  $\Delta V/V$  parameter remains below 40% throughout the simulation and it is somewhat lower than in the tholeiitic basalt case, the same negative consideration of Section 7.4.6 applies also in this case. To be quite explicit, the long-term sustainability of the process under consideration appears unlikely, owing to the dramatic reduction in the effective porosity of the aquifer. However, further investigations are warranted on the molar volume of silicate glasses and the role of water in these potentially interesting materials (*see* Section 7.5.1.2).

#### 7.6. Reaction path modelling of geological $CO_2$ sequestration in sedimentary basins

As shown above, ultramafic and mafic rocks have very high  $CO_2$  sequestration capacities. Nevertheless, they are not in general the best targets for geological  $CO_2$  disposal, owing to an unfavourable geological–hydrogeological framework.

First, rock matrix has low porosity and permeability, which determine low availability of water. The permeability of lavas develops mainly along fractures, cooling joints and discontinuous scoriaceous layers. The permeability of ophiolitic complexes is largely due to fracturing induced by the tectonic events that have determined the uplift of these bodies towards the surface and their emplacement at relatively shallow levels. Second, effective cap rocks are unlikely to be present at the top of tholeiitic basalt lava successions and ultramafic rocks variably affected by serpentinization, although relatively impervious layers may be locally present within these rock sequences. For instance, tuffs affected by argillic alteration may be associated with tholeiitic basalt lavas and clay-rich, deep-sea sediments usually occur together with ultramafic rocks in ophiolitic sequences. However, these impervious levels are randomly interposed into them, have limited lateral continuity and/or irregular shape due to the effects of tectonic events.

Favourable situations may be locally present, such as in southern Piedmont, Italy (Roberto Bencini, written communication). In this region,  $CO_2$  injection could be performed in the conglomerate layers, chiefly made up of serpentinitic clasts, which occur at the top of the serpentinities of the Voltri Group. Clogging of pore spaces should not be a major problem in these conglomerates, due to their high porosity and permeability. Besides, the overlying sedimentary rocks of the Piedmont Tertiary Basin should act as an efficient cap rock, preventing the escape of gaseous  $CO_2$  towards the surface. This fact must be absolutely avoided for obvious safety reasons! Further studies are needed to prove the feasibility of geological  $CO_2$  disposal in this region.

In contrast to mafic and ultramafic rocks, sedimentary basins are very promising targets for geological  $CO_2$  disposal. They have a favourable geological–hydrogeological framework, owing to the presence of both (i) high-porosity, high-permeability layers typically represented by sandstones and (ii) low-permeability layers of shales, potentially limiting the transfer of  $CO_2$  towards the surface. Besides, sedimentary basins host huge amounts of waters. At sufficiently deep levels, these waters typically have high salinities and, therefore, they are of no interest as sources of drinking water. The long residence times of these waters in the deep aquifers of sedimentary basins is an other requirement for the occurrence of carbonation reactions (mineral trapping) that are very slow processes, as already discussed above.

Sedimentary basins were often drilled at depth for the exploitation of oil and natural gas. Upon completion of the extraction of these valuable resources and perhaps after a first phase in which high-pressure  $CO_2$  is injected for enhanced oil recovery, abandoned wells can be used for geological  $CO_2$  disposal, without the need for drilling any new borehole. This is a very attractive situation, from the economic point of view, and money often makes things move.

For instance, 1 million ton of waste  $CO_2$  per year has been injected since September 1996 into the Utsira aquifer in the North Sea, to avoid a Norwegian tax of about 50 US\$ per ton of  $CO_2$  emitted offshore (Baklid et al., 1996; Torp and Gale, 2004). An other example is constituted by the Weyburn Oil Field in south central Saskatchewan, Canada which is currently the site of a large  $CO_2$  injection project for enhanced oil recovery (Emberley et al., 2004, 2005). Although this is not a geological  $CO_2$  sequestration project, approximately 30% of the injected  $CO_2$  will be trapped in the reservoir at the end of project activities (Gunter et al., 2000). Besides, the geological  $CO_2$  sequestration in sedimentary basins has been the subject of feasibility studies in The Netherlands (Lohuis, 1993) and in the Alberta Basin, Canada (Gunter et al., 1993; Bachu et al., 1994; Law and Bachu, 1996; Gunter et al., 1996, 1997, 2000).

Among the numerous modelling investigations on the geological  $CO_2$  sequestration in sedimentary basins, some selected examples are discussed in the following sections. Related experimental and field studies are treated in Sections 7.7 and 7.8.

#### 7.6.1. The glauconitic sandstone aquifer of the Alberta Sedimentary Basin

Reaction path modelling of geological  $CO_2$  sequestration in a glauconitic sandstone aquifer of the Alberta Sedimentary Basin, Western Canada was studied by Gunter et al. (1997, 2000) by using the PATHARC.94 computer code (Perkins and Gunter, 1995).

This glauconitic sandstone aquifer has an average porosity of 12% and its average mineral composition comprises 87% quartz, 2% K-feldspar, 1% plagioclase (which was taken into account in the geochemical simulation introducing the two endmember components albite and anorthite), 5% glauconite (which was hypothesized to be represented by annite in reaction path modelling), 2% kaolinite, 1% calcite, 1% dolomite and 1% siderite. The reactive surface area was assumed to be equal to the geometrical surface area and individual mineral grains were considered constant-volume sphere with an average radius of 0.05 mm. The average temperature of the aquifer is 54°C. The average chemical composition of formation water is (concentration in mg L<sup>-1</sup>) Na 28,800, K 690, Ca 2,970, Mg 578, HCO<sub>3</sub> 198, Cl 51,600 and SO<sub>4</sub> 366, with a pH of 7.2.

Gunter et al. (1997) simulated the injection of  $CO_2$  into the glauconitic sandstone aquifer at a constant  $P_{CO_2}$  of 260 bar. Substantial trapping of  $CO_2$  through precipitation of siderite (and subordinate calcite) and solubility trapping was predicted to take place in a time considerably smaller than the residence time of water in the sedimentary basin aquifer. The main occurring reaction was considered to be

 $3annite + 11CO_2 \rightarrow muscovite + 9siderite + 6quartz + H_2O + 2K^+ + 2 HCO_3^-.$ (7-50)

Gunter et al. (2000) chosen instead to dissolve a finite amount of CO<sub>2</sub>, 1 mol, in 1,000 g of water at the beginning of the simulation. In this way,  $P_{CO_2}$  decreases with  $\xi$ , simulating the conditions prevailing upon termination of CO<sub>2</sub> injection. They found that equilibrium is expected to be attained in hundreds of years with a drop of  $P_{CO_2}$  from 87 to 0.02 bar and the generation of 1 mol of siderite, 0.05 mol of calcite and 0.02 mol of dolomite. These figures confirm the relatively high CO<sub>2</sub>-trapping capacity of siliciclastic rocks containing Fe(II) minerals, such as annite.

However, these findings were questioned by Xu et al. (2000, 2004) who modelled the geological sequestration of CO<sub>2</sub> in the same glauconitic sandstone aquifer of the Alberta Sedimentary Basin by means of TOUGHREACT. They recognized that use of annite [KFe(II)<sub>3</sub>AlSi<sub>3</sub>O<sub>10</sub>(OH)<sub>2</sub>] as a proxy for glauconite [K<sub>0.75</sub>Mg<sub>0.25</sub>Fe(II)<sub>0.25</sub>Fe(III)<sub>1.25</sub>Al<sub>0.5</sub>Si<sub>3.75</sub> O<sub>10</sub>(OH)<sub>2</sub>] leads to overestimate the amount of precipitating siderite and, consequently, the CO<sub>2</sub>-sequestration capacity. Therefore they considered glauconite in reaction path modelling instead of annite. They also introduced oligoclase instead of albite plus anorthite and illite instead of muscovite. Note that the latter choice determines a moderate increase in the CO<sub>2</sub>-sequestration capacity, owing to the presence in illite of some Mg  $[K_{0.6}Mg_{0.25}Al_{1.8}(Al_{0.5}Si_{3.5}O_{10})(OH)_2]$  which is absent in muscovite. Besides, they assumed the presence of organic matter, namely type II kerogen.

To keep the description of precipitation kinetics to a simple level, backward rate constants were assumed to be one order of magnitude larger than forward (dissolution) rate constants. The total surface area was set at  $10 \text{ m}^2 \text{ dm}^{-3}$  of medium, corresponding to a surface of  $8.33 \times 10^5 \text{ cm}^2$  in contact with 1,000 g of water based on the average porosity of 12%. The surface area of secondary minerals was set to  $0.25 \text{ m}^2 \text{ dm}^{-3}$  throughout the simulation, apart from clay minerals, whose surface area was assumed to be one order of magnitude higher, for glauconite, and two orders of magnitude higher, for all the other clay minerals. Note that some of these phases are not initially present in the considered system, but they are considered to have an initial surface area different from zero. This arbitrary assumption (or an alternative one, e.g., instantaneous equilibrium as we assumed above) is needed to describe the precipitation of secondary minerals in a kinetic framework.

The initial aqueous solution was assumed to be a 1 M NaCl solution. The geological CO<sub>2</sub> sequestration was simulated by setting the CO<sub>2</sub> injection pressure at 260 bar and temperature at 54°C at all times. The main precipitating solid phases resulted to be illite, K-feldspar, siderite, and ankerite accompanied by minor amounts of dawsonite and dolomite. Carbon dioxide is mainly incorporated in ankerite and siderite. After ~15,000 years, the total CO<sub>2</sub> sequestration capacity (solubility trapping plus mineral trapping) increases close to linearly with time, attaining the value of 17 kg m<sup>-3</sup> of medium in 100,000 years. This corresponds to approximately 140 g kg<sup>-1</sup> water, a value which is attained much earlier in the cases of dunite, serpentinite, tholeiitic basalt and basaltic glass that were discussed above. This indicates that the sequestration capacity of glauconitic sandstones is much lower than those of previously considered lithotypes, in line with the smaller total contents of Mg-, Ca-, Fe(II)- and Na-bearing silicates. Porosity is expected to decrease very slightly with time from the initial value of 12% to 11.6% in 100,000 years.

A possible uncertainty relates to the exclusion of reduced sulphur species from geochemical modelling. Indeed, Xu and coworkers recognize that even low  $P_{\rm H_2S}$  would determine the production of pyrite, which could fix a significant proportion of Fe<sup>2+</sup>, thus decreasing the amount of precipitating siderite (and ankerite, but calcite and dolomite would form instead of it) and, consequently, the CO<sub>2</sub> sequestration capacity of the process.

# 7.6.2. The sediments of the Gulf Coast

After a preliminary study of Apps (1996), reaction path modelling of geological  $CO_2$  sequestration in the Gulf Coast sediments was investigated by Xu et al. (2000, 2004), again by means of TOUGHREACT.

These sediments have been considered to have an average porosity of 10% and to be made up of 49.7% (by volume) quartz, 1.8% kaolinite, 1.7% calcite, 0.85% illite, 2.6% organic matter (type II kerogen), 17.8% oligoclase, 7.4% K-feldspar, 3.5% Na-smectite, 2.7% clinochlore-14Å, 1.4% daphnite-14Å, 0.44% haematite. The total surface area was assumed to be 10 m<sup>2</sup> dm<sup>-3</sup> of medium, corresponding to a surface of  $1 \times 10^6$  cm<sup>2</sup> in contact with 1,000 g of water.

Again, a 1 M NaCl solution was taken into account as initial aqueous solution. The geological CO<sub>2</sub> sequestration was simulated by setting the CO<sub>2</sub> injection pressure at 260 bar and temperature at 80°C. The most important product solid phases resulted to be illite, daw-sonite and ankerite, whereas calcite and siderite initially precipitated and then re-dissolved. Carbon dioxide is mainly incorporated in dawsonite and ankerite. Precipitation of carbonate minerals determines an increase in the total CO<sub>2</sub> sequestration capacity up to a maximum of 90 kg m<sup>-3</sup> of medium (900 g kg<sup>-1</sup> water) after 100,000 years. Porosity decreases with time from the initial value of 10% to a value somewhat higher than 6% after 100,000 years. The sequestration capacity of these sediments is significantly higher than that of the glauconitic sandstones, mainly owing to the abundance of oligoclase and chlorite minerals.

The geochemical processes occurring in a sandstone–shale system representative of the Gulf Coast sediments upon injection of high-pressure  $CO_2$  at depths of ~2 km were investigated further by Xu et al. (2005). They performed numerical simulations of reactive fluid flow and geochemical transport by fully exploiting the capabilities of the computer code TOUGHREACT. Although the theoretical treatment of this subject goes beyond the aims of this book (the interested reader is referred to Lichtner et al., 1997), it is very interesting to give a look at the results of this type of advanced geochemical–hydrogeological modelling.

Xu et al. (2005) adopted a simplified one-dimensional model made up of a 10-mthick layer of shale overlying a layer of sandstone of the same thickness. The model grid (comprising all the blocks in this 1-D space) is finer in the shale near the sandstone/shale boundary, to appreciate the concentration gradients that are expected to develop in the shale, owing to its low permeability. In contrast, perfect mixing was assumed to occur in the sandstone layer, which was modelled by using a single block.

The same primary minerals listed above were considered to be present in the sandstone and in the shale layers, but in different amounts. Initial porosity was assumed to be 30% in the sandstone and 10% in the shale. Initial water chemistry was computed by reacting a 1 M NaCl solution with the primary minerals at 75°C under a  $P_{CO_2}$  of 0.01 bar. The obtained solution composition was taken as an initial condition in reactive transport modelling.

At the beginning of the reactive transport simulation,  $CO_2$  was added to the sandstone block to achieve a gas phase saturation of 0.5 and a  $P_{CO_2}$  of 201 bar, whereas no  $CO_2$ was added to the blocks of the shale layer, which were considered to be water saturated.

Changes with time in both water chemistry and the amounts of primary and secondary minerals along the sandstone–shale transect are computed by TOUGHREACT. For instance, the H<sup>+</sup> ion and aqueous CO<sub>2</sub> are found to diffuse from the sandstone to the shale, in line with the high  $P_{CO_2}$  imposed to the sandstone, which increases the H<sup>+</sup> activity of the aqueous phase. Initially carbonate mineral precipitation takes place chiefly in the sandstone, but afterwards it extends to the shale (though in lower amounts) owing to diffusive transport of CO<sub>2</sub> from the sandstone to the shale. Total dissolved Fe and Ca are expected to diffuse from the shale to the sandstone, since precipitation of siderite and ankerite causes a decrease in Fe<sup>2+</sup> and Ca<sup>2+</sup> concentrations in the sandstone. Precipitation of carbonate minerals (mainly dawsonite and ankerite) brings about a decrease in porosity from 30% to 24% after 100,000 years in the sandstone and rather complex changes in porosity in the shale.

The impacts of NO<sub>x</sub>, SO<sub>x</sub> and H<sub>2</sub>S (which are typically present together with CO<sub>2</sub> in coal-fired waste streams) on the geological disposal of CO<sub>2</sub> in the sands of the Frio Formation was investigated by Knauss et al. (2005, see also Knauss et al., 2001). They carried out reactive transport modelling by using the reactive transport simulator CRUNCH and showed that the presence of H<sub>2</sub>S should not interfere with CO<sub>2</sub> injection, whereas oxidation of small amounts of SO<sub>2</sub> could determine a significant decrease in pH and substantial differences in the process. This study has a very important applied component as the separation and compression of CO<sub>2</sub> is very expensive and represent the 75% of the total cost of the whole geological disposal process.

#### 7.6.3. The White Rim Sandstone

This case history was investigated by White et al. (2005) along a cross section beneath the Hunter Power Plant in Central Utah, emitting yearly  $15 \times 10^6$  tons of CO<sub>2</sub>. The Permian White Rims Sandstone with a thickness of 61–244 m, high porosity and permeability and lying at over 1 km depth beneath the power plant was selected as the best potential reservoir. It is overlied by some potential seals, represented by shale and evaporite levels. In the investigated area, the sedimentary sequence is a sort of gently dipping cuesta, constituting the western flank of the San Rafael asymmetrical anticline. Owing to the regional dip, the White Rims Sandstone crops out at the surface some 40 km to the SE, posing a possible problem of lateral migration and approach to the surface of the CO<sub>2</sub>charged aqueous solution after prolonged CO<sub>2</sub> disposal.

Particular attention was devoted to the sealing properties of potential cap rocks, which were measured through mercury injection porosimetry by White et al. (2005). As an example, it was found that the second most effective cap rock (Mancos Shale) could withhold a  $CO_2$  column of the order of 500–900 m in thickness.

Reactive transport modelling of  $CO_2$  injection in the White Rims Sandstone was carried out by using the ChemTOUGH code (White, 1995), which is based on the porous media multiphase mass and energy flow code TOUGH2 (Preuss, 1991).

The reservoir rock was assumed to be made up of quartz (77% by volume), kaolinite (2.25%), Na-smectite (2.25%), calcite (1.8%), anorthite (0.66%), albite (0.60%), K-feldspar (0.60%). Note that the complement to 100% is represented by porosity, approximately 15%. Dolomite, siderite and dawsonite are not initially present in the reservoir rock but were considered as possibly precipitating solid phases. Aquifer temperature was fixed at 54°C. Initial water chemistry was computed by reacting a 3 M NaCl brine with the reservoir minerals for 1,000 years.

In the simulation, injection of  $CO_2$  was terminated after 30 years. It was found that 1,000 years after the end of the injection period ~21% of the injected  $CO_2$  was trapped in carbonate minerals (calcite and dawsonite), ~52% was present underground as a separate gas phase or dissolved in groundwaters and ~17% had leaked to the ground surface, a process that was still going on.

#### 7.6.4. The shales of the North Sea

Gaus et al. (2005) carried out reaction path modelling (both in stoichiometric mode and in time frame) and reactive transport modelling to investigate the impact of high-pressure  $CO_2$  injection on the porosity of the cap rock at the Sleipner site, Norwegian part of the North Sea, where  $CO_2$  has been injected since 1996. The code PHREEQC v2.6 (Parkhurst and Appelo, 1996) was used to this purpose.

Carbon dioxide is currently injected in the sands of the Utsira Formation but it migrates upward, due to buoyancy effects, and accumulates below the overlying cap rock, represented by the shales of the 250-m-thick Nordland Group. Then  $CO_2$  dissolves into the formation water of the cap rock and diffuse upward into it. At Sleipner, the migration of injected  $CO_2$  is predicted and monitored by means of seismic surveys and reservoir modelling (Arts et al., 2004; Torp and Gale, 2004).

The mineral constituents of the cap rock considered in geochemical modelling are illite (24.7 wt.%), chalcedony (21.5%), kaolinite (18.0%), plagioclase (12.4%), a high-Fe-Mg smectite (8.8%), 14 Å-clinochlore (4.1%), pyrite (2.8%), K-feldspar (2.1%), siderite (1.6%) and calcite (1.0%). In three separate runs plagioclase was considered to be represented by either albite, or 50% albite plus 50% anorthite, or oligoclase ( $An_{20}$ ).

Initial porosity was set to 5%. Initial chemistry of formation water was computed equilibrating an aqueous solution containing 0.479 M chloride with cap rock minerals at 37°C.

Isothermal, isobaric injection of CO<sub>2</sub> was assumed at  $P_{CO_2}$  of 101 bar ( $f_{CO_2}$ =52 bar) and temperature of 37°C. In addition to primary minerals, dawsonite, disordered-dolomite and magnesite were selected as possible secondary minerals, defining suitable rate constants and reactive surface areas different from zero (these are arbitrary but necessary assumptions, as already noted above).

Reaction path modelling in time frame shows that the main secondary carbonate mineral is calcite accompanied by a limited amount of dawsonite in all the three runs with different plagioclase composition. Small changes in porosity (<0.05 %) are calculated for the albite case, whereas a certain decrease in porosity (up to 2.8% in the first meter) is predicted for the albite plus anorthite case. This is attributed to the dissolution of anorthite and the subsequent precipitation of calcite and kaolinite by Gaus et al. (2005). It must be noted that the volume of the reaction:

$$CaAl_{2}Si_{2}O_{8(s)} + CO_{2(g)} + 2H_{2}O \rightarrow CaCO_{3(s)} + Al_{2}Si_{2}O_{5}(OH)_{4(s)},$$
 (7-51)

 $\Delta V_r$ , is 35.7 cm<sup>3</sup> (per mole of anorthite, that is referring to the reaction as it is written) and the corresponding  $\Delta V/V$  parameter is 35.4% (*see* equation (5-125)), if consideration is restricted to the solid phases only, as was done in the previous sections of this chapter and in Section 5.5. The  $\Delta V_r$  reduces to -0.37 cm<sup>3</sup> and the  $\Delta V/V$  parameter becomes -0.27%if water is also included in calculations as evidently assumed by Gaus et al. (2005). In my opinion, this assumption is reasonable as long as water is forced to remain in the considered system, which is rather unrealistic in natural systems. Therefore, porosity changes have to be taken with some caution. In spite of this divergent opinion, the study by Gaus et al. (2005) is very important as it focuses the attention on the integrity of the cap rock, which is a very important issue for the safe geological disposal of  $CO_2$ .

The processes relevant for the geological sequestration of  $CO_2$  at the Sleipner site were also investigated in a very interesting study by Johnson et al. (2001), who used an integrated toolbox comprising NUFT, a software package for the numerical simulation of multiphase/multicomponent flow and reactive transport (Nitao, 1998), GEMBOCHS, a thermodynamic/kinetic database and dedicated software library (Johnson and Lundeen, 1994a, b, 1995), and the graphic utility Xtool (Daveler, 1998). The Sleipner environment is represented by a spatial domain comprising an Utsira-like 200-m-thick saline aquifer confined by a 25-m-thick shale cap-rock, which in turn is overlain by a 25-m-thick aquifer. Three distinct aquifer scenarios were considered, differing for the absence or presence of intra-aquifer shales of different lateral continuity and permeability (the latter parameter was varied to simulate the absence or presence of microfractures). A 20-year-long time frame was adopted, comprising a prograde phase (in which the injection rate is kept constant at 10,000 ton  $CO_2$ /year), a transition phase (in which the injection rate is decreased to 0 in 3 months) and a retrograde phase (in which the injection rate is maintained at 0 for 9.75 years).

The study by Johnson et al. (2001) showed that the migration of immiscible  $CO_2$ and the relative importance of solubility and mineral trapping processes are chiefly controlled by the distribution of intra-aquifer permeability, whereas the long-term storage security is mainly affected by cap-rock integrity. Intra-aquifer shales retard the vertical migration of the immiscible  $CO_2$  plume and promote its lateral expansion (in agreement with the results of seismic surveys; Torp and Gale, 2004), thus increasing the spatial extent of plume-aquifer interaction and the efficacy of intraplume solubility and mineral trapping processes. The 25-m-thick shale cap-rock resulted to be an efficacious barrier prohibiting the vertical migration of the immiscible  $CO_2$  plume (again, in agreement with seismic evidence; Torp and Gale, 2004). In the relatively short interval of time considered in the simulation,  $\sim 85\%$  of the injected CO<sub>2</sub> constitutes an immiscible supercritical fluid phase, ~15% dissolves in the aqueous solution and <1% precipitates as carbonate minerals, in the form of dawsonite, magnesite/siderite and calcite. However, mineral trapping is enhanced during the retrograde phase and has very important consequences, in that it limits the lateral extent of plume migration and increases the cap-rock integrity, through carbonate sealing of microfracture porosity.

#### 7.6.5. The carbonate rocks of the Alberta sedimentary basin

Similar to the glauconitic sandstone case (*see* Section 7.6.1), Gunter et al. (2000) modelled the effects of the dissolution of a finite amount of  $CO_2$  (1 mol in 1,000 g of water) also on the Nisku carbonate aquifer of the Alberta Sedimentary Basin. These carbonate rocks consist of prevailing dolomite and/or calcite (both containing traces of Fe), possibly accompanied by small amounts (<1% in general) of anhydrite, illite, pyrite and quartz. Silicate and sulphide minerals were excluded from the simulation, which was carried out for a system comprising 14,000 g of dolomite, 1,100 g of calcite, 262 g of siderite and 1,000 g of water.

Equilibrium is predicted to be achieved in less than 1 day in the carbonate aquifer, owing to the occurrence of fast reactions, with a small reduction in  $P_{\rm CO2}$  from 87 to 86 bar, the dissolution of small amounts of calcite (0.06 mol) and siderite (10<sup>-5</sup> mol) and the precipitation of 0.02 mol of dolomite.

# 7.7. Water–rock reactions during geological CO<sub>2</sub> sequestration: the experimental evidence

#### 7.7.1. Laboratory experiments on rocks from sedimentary basins

Precipitation of clay minerals and carbonate minerals as a consequence of dissolution of primary silicates and Al-silicates was generally not observed in laboratory experiments aimed at evaluating the interactions between  $CO_2$ , formation water and rocks from sedimentary basins. This is due to the slow kinetics of these processes. Exceptions are represented by the recent experimental works by Kaszuba et al. (2003, 2005).

Pearce et al. (1996) and Rochelle et al. (1996) injected supercritical  $CO_2$ , at pressures of 90 and 200 bar, into a sandstone at temperatures of 105° and 80°C, for periods of 1 and 8 months. Significant alteration of pre-existing calcite and dolomite was observed, as well as dissolution of anhydrite accompanied by precipitation of calcite and possible corrosion of detrital feldspars with precipitation of Na-smectite, although the evidence supporting this last process was not conclusive.

Gunter et al. (1997) performed 1-month-long experimental tests at 105°C and 90 bar of  $P_{CO_2}$  on glauconitic sandstones from the Alberta sedimentary basin. Under these conditions, only a small amount of CO<sub>2</sub> was trapped through reactions with Al-silicate minerals, whereas dissolution of mineral carbonates took place.

Sass et al. (2001) reacted anorthite and glauconite with  $CO_2$  and a Na–Ca–Cl synthetic brine (containing other major components and several traces) at 50°C, 41 bar  $P_{CO_2}$  and 150°C, 138 bar  $P_{CO_2}$ , under a total pressure of 138 bar. The water–rock–gas interaction experiments were carried out in vessels placed on a shaker table or rocker for a duration of approximately 1 month. Anorthite incorporated some Na in the lattice and released a corresponding amount of Ca to the aqueous solution but precipitation of carbonate minerals did not occurred. In the experimental runs with glauconite, the Na content of the solid phase increased, K, Fe and Si decreased, whereas Al, Mg and Ca did not change. Also in this case, no carbonate mineral formed as a reaction product.

Kaszuba et al. (2003, 2005) simulated a brine–aquifer–aquitard system in a flexible cell hydrothermal reactor. A synthetic arkose, assumed to represent the aquifer, was prepared by mixing equal amounts of quartz, oligoclase  $(An_{17-21})$  and microcline  $(Or_{91-97})$  showing perthitic lamellae  $(Or_{3-9})$  and adding some biotite. An argillaceous shale, made up of clay minerals (65%, by volume) quartz (27%), feldspar (5%), chlorite (2%, possibly including some kaolinite) and trace amounts of framboidal pyrite, was used for the aquitard. Kaszuba et al. (2003) used a synthetic brine containing Na<sup>+</sup>, Mg<sup>2+</sup>, K<sup>+</sup> and Cl<sup>-</sup> as prevailing constituents whereas a pure Na–Cl brine was employed by Kaszuba et al. (2005), both with ionic strength of 5.5 mol kg<sup>-1</sup>. In a first step, the system was maintained

at 200°C, 200 bar for several days (59 in the 2003 run, 32 in the 2005 run) to attain steady state. Then  $CO_2$  was added to the reactor. In a second step, the system was kept again at 200°C, 200 bar, for 80 days in the 2003 run and 77 days in the 2005 run. Kaszuba et al. (2005) performed also a reference experiment on the same system, at the same P, T conditions and for the same time, without  $CO_2$  injection.

Kaszuba et al. (2003) documented (i) etching of potassic lamellae in perthitic microcline with respect to sodic lamellae; (ii) growth of clay minerals on oligoclase; (iii) coatings of magnesite associated with clay minerals on biotite grains; (iv) growth of analcime with skeletal habit.

Kaszuba et al. (2005) observed the (i) presence of corroded magnesite and euhedral siderite, suggesting the early occurrence of magnesite precipitation followed by magnesite dissolution and siderite precipitation; (ii) growth of clay minerals and analcime. Analcime was found to be more abundant in the  $CO_2$ -free experiment where it shows an euhedral texture. In contrast, analcime crystals with skeletal texture grew in the  $CO_2$ -brine-rock experiment, as also found by Kaszuba et al. (2003).

Significant variations were also detected in brine chemistry. In particular, the significant increase in the concentrations of several components, initially absent in the Na–Cl brine used by Kaszuba et al. (2005), undoubtedly indicate the occurrence of rock dissolution through reaction with  $CO_2$  and brine. All in all, these observations are consistent with the occurrence of dissolution of primary silicate minerals and precipitation of both carbonates and silicates.

Kaszuba et al. (2005) underscored that "precipitation of the mixed hydroxyl carbonate mineral dawsonite (NaAlCO<sub>3</sub>(OH)<sub>2</sub>), predicted as a stable carbonate phase in reactive transport models of carbon sequestration (Johnson et al., 2002; Knauss et al., 2002) was not observed in this study." They estimated that the aqueous solution was probably undersaturated with dawsonite, although the thermodynamic affinity with respect to this mineral and other Al-bearing solid phases could not be computed because the Al concentration in the brine was below detection limit throughout the CO<sub>2</sub>–brine–rock experiment. To try to elucidate this matter, it is worth taking into account the equilibrium reaction between dawsonite and analcime, an ubiquitous phase in the experiments of Kaszuba and coworkers:

$$Na_{0.96}Al_{0.96}Si_{2.04}O_{6} \cdot H_{2}O + 0.96 CO_{2(g)} \Leftrightarrow 0.96 NaAlCO_{3}(OH)_{2} + 2.04 SiO_{2} + 0.04 H_{2}O.$$
(7-52)

Assuming that the activity of aqueous SiO<sub>2</sub> is fixed by equilibrium with chalcedony and the activity of water and solid phases are equal to 1, it results  $\log f_{CO_2} = -\log K_{7-52}$  /-0.96. Using the thermodynamic equilibrium constants of EQ3/6, it turns out that at 200°C, dawsonite is stable for  $f_{CO_2} > 7.8$  kbar and analcime is stable below this  $f_{CO_2}$  value. Since CO<sub>2</sub> fugacities were much lower than 7.8 kbar in the experiments of Kaszuba and coworkers, the absence of dawsonite as a solid product phase is not surprising. We may also note that the  $f_{CO_2}$  of analcime–dawsonite–chalcedony equilibrium coexistence is 0.23 bar at 60°C, 7.2 bar at 100°C and 296 bar at 150°C. In spite of possible uncertainties on the thermodynamic data of dawsonite, these figures confirm that dawsonite likes cool environments, in line with what was inferred in Section 5.1.6 on the albite-dawsonite-quartz coexistence. Therefore, high temperature (200°C) has certainly enhanced the kinetics of dissolution/precipitation reactions, but has also stabilized analcime at the expense of dawsonite, in the experiments by Kaszuba and coworkers.

#### 7.7.2. Laboratory experiments on forsterite and serpentine

The carbonation reactions of forsterite and serpentine have been the subject of several studies, after Lackner et al. (1995, 1997) underscored that these Mg silicates obtained from ultramafic rocks are abundant and thermodynamically convenient primary minerals for mineral carbonation.

McKelvy et al. (2001) showed that dehydroxylated samples of serpentine (metaserpentine), obtained through heat treatment, undergo carbonation more readily than untreated samples. The optimum temperature of heat treatment was found to be 600–650°C by O'Connor et al. (2001), with little increase in reactivity upon heat treatment at higher temperatures. O'Connor et al. (2001) investigated also the mineral carbonation of olivine and meta-serpentine in aqueous slurries and found optimum results at 155°C under a  $P_{CO_2}$  of 185 bar for a 0.64 M NaHCO<sub>3</sub> and 1.0 M NaCl aqueous solution containing 15% of solids. Under these conditions a 78% conversion of the Mg-silicates to carbonate was attained in 30 min.

Guthrie et al. (2001) carried out experiments of serpentine, meta-serpentine and olivine carbonation in a stirred reaction vessel typically containing ~0.8 L of water plus solids and ~1 L of supercritical  $CO_2$  at temperatures up to 185°C and pressures up to ~160 bar. They proposed a geochemical model assuming (i) occurrence of dissolution and precipitation processes in the aqueous phase and not in the supercritical  $CO_2$  phase and (ii) occurrence of carbonation via dissolution–precipitation and not through direct carbonation of the Mg-silicate. The geochemical model comprises four steps: (i) dissolution of  $CO_2$  into the aqueous phase and of H<sub>2</sub>O into the  $CO_2$ -rich phase; (ii) release of Mg<sup>2+</sup> ion to the aqueous phase, which is considered to be the rate limiting step; (iii) speciation reactions in the aqueous phase; (iv) precipitation of magnesium carbonate. Magnesite resulted to be the most typical product, but hydromagnesite was observed in several runs upon either alteration of fluid chemistry or  $CO_2$  venting before quenching of the reaction vessel.

Based on the findings of O'Connor et al. (2001), Wolf et al. (2004) performed the carbonation of meta-serpentine in a solution 0.64 M NaHCO<sub>3</sub> and 1.0 M NaCl under a  $P_{CO_2}$  of 150 bar, by using a suitably implemented externally-controlled microreaction system. This allows *in situ* monitoring of the reaction process by synchrotron X-ray diffraction and Raman spectroscopy. By progressively heating the system, they were able to detect the first X-ray reflections of magnesite at 150°C, which is the temperature at which carbonation reaction begins. Further carbonation of meta-serpentine to magnesite occurred upon continuous heating to 180°C. However, no reflection caused by Si-bearing minerals was detected, indicating that SiO<sub>2</sub>-like products are highly disordered.

Further studies of this type were carried out by McKelvy et al. (2004), who used *in situ* synchrotron X-ray diffraction to investigate the reactivity of meta-serpentine (obtained

through heat activation of lizardite at different temperatures) under the optimal conditions of carbonation established by O'Connor et al. (2001). Trace carbonation was observed at 120°C for the meta-serpentine obtained at 580°C, but further heating did not cause any further carbonation, indicating that the meta-serpentine was only partially activated. Substantial carbonation was found at 125°C for the 610°C meta-serpentine. The best results were obtained for the 650 and 720°C meta-serpentine in terms of progress and onset temperature of the carbonation reaction, which resulted to be 100 and 105°C, respectively. Besides, McKelvy et al. (2004) investigated the structure and composition of meta-serpentine by using TGA/DTA, XRD and FTIR analyses. Longer-range ordering, possibly due to doubling of the basal spacing from the characteristic value of 7.3 Å (*see* Section 5.3.2.1) to 14.7 Å, was observed in the material treated at 580°C, whereas the crystalline features disappear at 650°C indicating formation of amorphous meta-serpentine.

Schulze et al. (2004) used XPS and TEM to investigate the surface of ground lizardite, untreated, heat-treated at  $630^{\circ}$ C in a CO<sub>2</sub> flux for 3 h and heat-treated and reacted with a Na–Cl–HCO<sub>3</sub> solution and CO<sub>2</sub> at 155°C for 2 h, thus achieving a 22% conversion to carbonate. The latter treatment determined the formation of a 40–100 nm surface layer constituted by amorphous silica containing Mg. According to Schulze et al. (2004), the rate-limiting step of lizardite carbonation is the diffusion of Mg<sup>2+</sup> ion from the amorphous phase.

Since heat treatment is a rather expensive operation, a different approach based on the physical activation of serpentine at low temperature coupled with a pH swing process was investigated by Park and Fan (2004). Fluidation of a serpentine slurry with 2 mm glass beads resulted to be the most effective method for removing the silica-rich surface layer and promoting further serpentine dissolution.

Giammar et al. (2005) investigated forsterite dissolution and magnesite precipitation through a series of batch experiments. Forsterite dissolution tests were carried out in dilute solutions (I 0.003 to 0.018 mol kg<sup>-1</sup>) at 30 and 95°C under initial  $P_{CO_2}$  of 1 or 100 bar, whereas magnesite precipitation experiments were run in MgCl<sub>2</sub>–NaHCO<sub>3</sub> solutions (I 0.07 to 1.25 mol kg<sup>-1</sup>) under initial  $P_{CO_2}$  of 100 bar. Several analytical methods (elemental analysis of the aqueous solution, SEM, FTIR and XRD for the solids) and geochemical modelling were used to elucidate reaction products and progress.

The extent of forsterite dissolution was found to increase with increasing  $P_{CO_2}$  and temperature. The release of Mg and Si during forsterite dissolution was stoichiometric, but the concentration of dissolved Si was limited by saturation with respect to amorphous silica. Even though the aqueous phase attained conditions of oversaturation with respect to magnesite (saturation index up to 0.25) during forsterite dissolution experiments, MgCO<sub>3</sub> did not precipitated.

Magnesite precipitation occurred in ad hoc experiments designed to estimate the influence of suspended solid particles (of either forsterite or magnesite), time and saturation index with respect to magnesite. The composition of the aqueous solution, MgCl<sub>2</sub>–NaHCO<sub>3</sub>, was selected to simulate extensive forsterite dissolution. Magnesite precipitation was found to be controlled by nucleation, which requires a critical saturation index between 0.25 and 1.14 at 95°C and  $P_{CO_2}$  of 100 bar. Magnesite seeds accelerate magnesite precipitation, whereas forsterite particles do not, although magnesite forms on their surfaces.

# 7.8. Water-rock reactions during geological CO<sub>2</sub> sequestration: the field evidence

Emberley et al. (2004, 2005) evaluated the gas-water-rock interactions that occurred upon miscible  $CO_2$  injection into the oil reservoir of Weyburn, Saskatchewan (Canada) for enhanced oil recovery (EOR). Oil is hosted into the carbonate rocks known as Midale beds, which are sealed by an anhydrite cap. The reservoir has an average temperature of 60°C and initial pressure was 146 bar on average. The  $CO_2$  is generated by the Great Plains Synfuel coal gasification plant in Beulah, North Dakota (USA) and is transported at Weyburn by a 320-km-long pipeline. Note that  $CO_2$  is injected together with water to retard  $CO_2$  mobility. Since  $CO_2$  is more soluble in oil than in water, it is released from water to oil. Owing to absorption of  $CO_2$ , oil experiences a decrease in viscosity and an increase in mobility. Besides, the low pH of the  $CO_2$ -rich aqueous solution can cause dissolution of carbonate minerals and increase in reservoir permeability. Although the aim of the miscibile  $CO_2$  injection project is EOR, there are also important environmental aspects since up to 30% of the injected  $CO_2$  is expected to be trapped in the reservoir.

To evaluate gas-water-rock interactions, a baseline fluid sampling was carried out before the onset of the CO<sub>2</sub> injection project and a second sampling survey (known as Monitoring-1) was performed 7 months after this date, when a total of approximately  $150.6 \times 10^6$  standard m<sup>3</sup> of CO<sub>2</sub> (corresponding to ~3 × 10<sup>5</sup> tons) were disposed in the oil reservoir. Data on water and gas chemistry and isotope compositions were obtained in both surveys.

The samples collected during Monitoring-1 differ from the baseline data for lower pH values with differences of 0.5–0.6 pH units, higher alkalinity which has doubled compared to the baseline values (although a significant fraction of titration alkalinity is due to dissolved sulphide), higher Ca concentrations and higher  $\delta^{13}$ C values of dissolved CO<sub>2</sub> by one to some  $\%_0$  units. All these differences were attributed to: (i) dissolution of the injected CO<sub>2</sub> in the aqueous phase as both carbonic acid and bicarbonate ion and (ii) concurrent dissolution of carbonate minerals, chiefly calcite and dolomite.

Nevertheless other chemical changes (e.g., in chloride and sulphate concentrations) also took place, indicating possible flushing of waters from other geological formations to the Midale beds and entrance in the monitored wells. Therefore, the overall picture is probably somewhat more complicated than that of simple  $CO_2$  absorption in the aqueous phase and consequent dissolution of carbonate minerals. Data obtained during subsequent monitoring phases add further complexities to the overall picture rather than clarifying it.

## 7.9. The need for a synergistic approach

Based on what was discussed in previous sections, it is quite evident that the reactions of dissolution of primary silicates and precipitation of secondary minerals, including carbonates, proceed to relatively limited extents in some laboratory experiments and field tests and do not occur at all in others. Geochemical modelling generally predicts that comparatively long lapses of time, in the order of hundreds, thousands, or even tens of thousands years, are needed for the significant progress of carbonation reactions. Therefore, there is no contrast between the results of geochemical modelling and those of laboratory experiments and field tests. Experiments and tests are entirely compatible with the computer simulation of geochemical processes and, in our opinion, a synergistic use of the two approaches is recommendable, as already done in some high-quality studies (e.g., Gunter et al., 1997 and Giammar et al., 2005 among the others).

Geochemical models are very attractive as it takes a comparatively short lapse of time to obtain results that would be achieved after much longer intervals of time, sometimes not accessible to humans, by means of real experiments either in the laboratory or in the field. However, users of geochemical models must be aware of possible dangers.

One of these is to rely too much on computer experiments or to expect too much from them. This was clearly discussed by Giggenbach (1984), by means of the following words.

As the reality of these models increases there is a danger, however, that their complexity and intractability will soon rival those of the natural systems studied. The emerging gap between human and machine understanding of these systems is illustrated by a remark by Reed (1982) in which he stresses the utility of the computer approach "when one considers how difficult it would be to apply intuitive reasoning and partial calculations to determine the course of a water-rock reaction. The large number of unknowns and the non-linearity of the equations make it difficult to understand the complexly intertwined processes." Taking into account the fact that intuitive thinking, a prerogative of human mind, is the basis of all scientific progress, relegating the process of understanding our environment to machines may eventually stifle rather than further scientific development.

Besides, it must be noticed that the validity of the results of computer experiments depends chiefly on how we have set up the model at the beginning of the simulation. No good answer can be expected from an ill-posed question whereas a good answer can be obtained if we succeed to formulate a correct question. Again, *setting-up a valid model requires an at least general, but conceptually accurate understanding of the processes active*, as underscored by Giggenbach (1984).

The outputs of computer experiments are also influenced by the quality of the thermodynamic and kinetic data, the validity of the used algorithms and many other "details". For instance, the application of geochemical models at high pressures and salinities is not straightforward, as discussed by Allen et al. (2005). See also Section 7.3.3 for the salinity effects.

Last but not least, presently available models are far from perfect as already discussed above and summarized in the next section.

# 7.9.1. Current limitations and future developments of reaction path modelling

The chemical reactions between minerals, the aqueous solution and  $CO_2$  occurring upon high-pressure  $CO_2$  injection are very complex and the theoretical framework needed

for their description is not fully incorporated in existing computer codes. Some of these deficiencies are summarized below.

- (1) Available codes incorporate at best the extended (B-dot) Debye–Hückel equation to compute *individual ion activity coefficients*,  $\gamma_i$ , although its application to brines is not advisable. Ideally, one should use the Pitzer's approach to compute the  $\gamma_i$  of solute species, but needed parameters are generally not available at high temperature and are also lacking at all temperatures for pivotal aqueous species such as dissolved SiO<sub>2</sub>, Al<sup>3+</sup> and related complex species.
- (2) The thermodynamic properties of some mineral phases of interest are unknown or poorly known. This is especially true for clay minerals, which are usually described in terms of idealized, limiting compositions.
- (3) The kinetic database is even less complete than the thermodynamic database and it is difficult to fill this gap as no general laws of overall reaction kinetics exist in parallel with the general laws of thermodynamics, and no necessary genetic relationship with which to connect kinetic species to thermodynamic species is known (Sposito, 1994).
- (4) In particular, the *influence of CO*<sub>2</sub> on the dissolution rate has been investigated for a limited number of mineral phases up to 1 bar  $P_{CO2}$  and only for labradorite, calcite, dolomite and magnesite up to 50–68 bar  $P_{CO2}$ . Although available data indicate that the effect of CO<sub>2</sub> is that of controlling the pH of the aqueous solution rather than participating directly to the dissolution mechanism of solid phases, this matter deserves further consideration.
- (5) The processes occurring during the *precipitation of solid phases* (e.g., nucleation and crystal growth) are not described adequately. The backward (precipitation) rate laws incorporated in available codes require the knowledge of the initial surface area, but this is zero for a phase that is not present in the system at the beginning of the simulation. Adoption of a value different from zero or assumption of instantaneous precipitation are two possible choices for circumventing this difficulty but this way to proceed is far from satisfactory.
- (6) Possible variations of the *reactive surface areas* with reaction progress, due to armouring of mineral phases by other phases and/or other reasons, cannot be described.
- (7) In natural systems, the formation of a secondary phase rather than another often depends on their different kinetics of growth, which lead to *different reaction paths*. This kind of problem cannot be tackled by means of available codes.

For all these deficiencies, the geochemical evolution of the systems of interest upon high-pressure  $CO_2$  injection cannot be predicted with sufficient accuracy by means of computer experiments at present. However, geochemical modelling is the only available tool for the evaluation of long-term geochemical processes, including those of interest for geological  $CO_2$  sequestration. In spite of possible improvements in the experimental approach, there is little hope to extend their duration to multiples of the life time of humans even from a very optimistic point of view. In contrast, computer experiments can be improved in future to understand the long-term evolution of the systems of interest.

# 7.10. A final note

I tried to be consistent throughout this book in the use of symbols and unit of measurements, and, even more important, in the adoption of chemical species. For what concerns the symbols, I am aware that I was not able to fulfil my initial purposes and, more important, the expectations of possible readers.

For what concerns the units, I preferred to use the bar and sometimes the atmosphere (1 atm = 1.01325 bar) instead of the recommended unit, i.e. the Pascal (1 MPa =  $1 \times 10^6$  Pa = 10 bar), as I think that bar and atm are more immediate and related to the real world than Pa. I expressed the thermodynamic affinity A in J mol<sup>-1</sup> (or kJ mol<sup>-1</sup>) in Chapter 6 but I used cal mol<sup>-1</sup> (or kcal mol<sup>-1</sup>) in Chapter 7, as this is the unit adopted by EQ3/6. After all, one has to be able to do conversions. This is especially important for energies, as they are also involved in dietetic suggestions. Energy conversions are done by multiplying the A value expressed in cal mol<sup>-1</sup> (or kcal mol<sup>-1</sup>) times 4.184 to obtain the corresponding value in J mol<sup>-1</sup> (or kJ mol<sup>-1</sup>).

Unfortunately, the adoption of chemical species is not always the same in the geochemical literature and, in particular, there is some confusion on the species adopted to describe neutral dissolved CO<sub>2</sub> species. As already recalled (see Chapter 2), both true dissolved molecular carbon dioxide, CO<sub>2</sub>°, and true carbonic acid, H<sub>2</sub>CO<sub>3</sub>°, coexist in the aqueous solution and, at equilibrium, the  $m_{\rm H_2CO_3^\circ}/m_{\rm CO_2^\circ}$  ratio is 0.0026 at 25°C, 1 bar (Sposito, 1994). However, in the geochemical literature the equilibrium between CO<sub>2</sub>° and H<sub>2</sub>CO<sub>3</sub>° is usually neglected and these two species are considered together in a lumped parameter. This is called apparent carbonic acid, H<sub>2</sub>CO<sub>3</sub>\*, by some authors and aqueous carbon dioxide, CO<sub>2(aq)</sub>, by others.

In this book, I have decided to follow the approach of each author, although this brings about some inconsistencies, of which the careful reader has already been aware at this point.

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