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Unsteady Heat Conduction Study During Underground Coal Gasification For Constructing an In-situ Pyrolysis Project.

دراسة التوصيل الحراري أثناء تصعيد الغازات من الفحم الكائن في باطن الأرض من أجل تصعيد مشروع تحلل حراري للفحم في موضحة

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ملخص البحث

في هذا البحث تم دراسة التوصيل الحراري الغير مستقر في طبقات الصخور و الفحم في باطن الأرض أثناء تصعيد الغازات من الفحم و ذلك عن طريق استخدام مفاعلات نووية كروية الشكل و هذه المصادر بها درجات حرارة عالية و يتم إدخالها في طبقات الصخور الموجودة بين كل طبقتي فحم أو في طبقات الفحم مباشرة وكذلك تم دراسة تغيير درجات الحرارة داخل طبقات الفحم وكذلك تأثير تغيير كمية الفحم المسمخن و المحلل حراريا و تحديد درجة الحرارة اللازمة لحدوث عملية التحلل الحراري حيث ينتج من تسخين الفحم انبعاث كمية لا بأس بها من الغازات تتكون أساسا من غاز الهيدروجين، الميثان و التي يمكن جمعها و ضخها خلال خطوط أنابيب لاستخدامها في أغراض صناعية مختلفة و في هذا البحث تم استخدام شكل كروي درجة حرارة سطحة 1000°م، 1500°م، 1750°م، 2000°م موضوع بشكل متماثل في طبقة الصخور و الفحم ولكي يتم حساب توزيع درجة الحرارة كدالة في الزمن فقد تم وضع وحل المعادلات التفاضلية التي تحكم عملية التوصيل الحراري الغير مستقر في طبقات الفحم و الصخور عدديا باستخدام طريقة

"The discretization equation (control-volume formulation)" . و قد أظهرت النتائج النظرية و العملية أن شكل و توزيع درجات الحرارة و كذلك درجة الحرارة القصوى التي يمكن الوصول إليها في طبقات الفحم يعتمد على عنصر الزمن و كذلك درجة حرارة سطح مفاعل التسخين بالإضافة إلى أن درجة الحرارة اللازمة لانبعاث الغازات من الفحم هي 400 درجة مئوية كما أن سرعة تسخين الفحم تزداد في الأيام الأولى للتسخين ثم تبدأ في التناقص حتى يتم الوصول إلى حالة الاستقرار والتي عندها سرعة التسخين تقترب من الصفر بعد ذلك و يختلف زمن الوصول إلى حالة الاستقرار مع اختلاف طريقة التسخين و كذلك درجة حرارة سطح مفاعل التسخين.

Abstract

Underground coal gasification (UCG) is a potential coal utilization technology, receiving renewed interest around the world. In the last five years there has been increasing interest in the world energy and mining industry for the development of a new operating underground coal gasification systems to generate synthetic gas from underground coal seams. In this paper, the problem of unsteady heat conduction in the underground coal and rock layers during underground gasification is considered. The heating of coal results in releasing considerable amount of gases mainly hydrogen and methane which would be collected and pumped through pipelines for the use in different industrial applications. In the present analysis, a spherical type heat source having a surface temperature of (1000°C, 1500°C, 1750°C and 2000°C) is used for heating the coal layer for achieving the pyrolysis process. The differential equations describing the unsteady heat conduction in both coal and rock layers are solved numerically using the discretization method (control-volume formulation). A computer program is constructed for calculating the temperature fields and the gasified volumes of mineral coal as well as the heating velocity in the coal seams. In the experimental part, a model of underground coal layer is constructed. To validate the numerical model, an electrical spherical heat source is inserted directly into the coal layer for heating. The temperature distribution in coal layer is measured in different locations using the

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thermocouples. The results show that the temperature profiles and the maximum attainable temperature in the coal seams depend on the time and the surface temperature of the heating reactor. Furthermore, the gasified volume of coal increases with the increase of both the surface temperature of the heating source and the running time. It is found also that the heating velocity in the underground coal lies between the mean value (2 to 10^{-3} K/min.) and increases with the time to reach a maximum value, after this it decreases to approach zero at the steady-state conditions. This study indicates that geometry, position, surface temperature and distribution of heating sources are very important parameters which should be carefully chosen for realizing an efficient pyrolysis of underground coal and avoiding thereby the combustion of a part of the coal seams. The analysis indicated that the best temperature is between (700 - 800°C) at which the high efficiency of pyrolysis process, the analysis might be useful in constructing an in-situ pyrolysis project for utilizing the underground coal seams.

NOMENCLATURE

T	temperature($^{\circ}\text{C}$)
T_i	Initial temperature
T_o	Old temperature
t	time
ρ	density kg/m^3
c	specific heat ($\text{kJ}/\text{Kg.K}$)
	thermal conductivity($\text{W}/\text{m.K}$)
R	radius (m) of heat source
f	a weighting factor
x, y, z	coordinate
Δt	time step
Δx	x-direction width of the control volume
δx	x- direction desistance between two adjacent grid points
$\Delta y, \delta y$	similar to $\Delta x, \delta x$
$\Delta z, \delta z$	similar to $\Delta x, \delta x$
(Φ)	general dependent variable
ρ	central grid point under consideration
α	thermal diffusivity (cm^2/s)

1.1 INTRODUCTION

The term "coal" used to describe a variety of fossilized plant materials, but no two coals are exactly alike. Heating value, ash melting temperature, sulfur and other impurities, mechanical strength, and many other chemical and physical properties must be considered when matching specific coal to a particular application. Gaseous fuel are among the most sought - after raw materials and energy sources in the energy

economy. Although the gasification of coal was practiced as long ago as the 19th century and was developed to an advanced stage of technology in the first half of the present century, especially in Europe, nowadays it is of importance only in a few countries with special economic conditions. Three main routes can be followed for producing gas from coal on a large industrial scale. Processes involving pyrolysis are the most important, both quantitatively and economically. During pyrolysis, the coal volatile constituents are cracked to give gaseous and liquid products, which can be separated and treated to provide marketable by-products.[1]. In the primary-metal industries coal is used to produce coke for blast furnaces and foundries and for power generation in steel and rolling mills. Significant amounts of coal are also used for power generation in the food, paper, chemical and ceramic industries. Pyrolysis typically occurs under pressure and operating temperature above 430°C (EX-situ Remediation technology)[6]. Pyrolysis processes can be divided into two groups: low temperature and high temperature. The product of pyrolysis processes

can be controlled by temperature and the rate of material heating. At high temperature, the major product is gas, while at the low temperature The major product is tar or heavy oil. High heating rates minimize char formation by breaking down higher molecular species into gases products. The Laminar Energy Research Center [LERC] conducted an underground coal gasification experiment in sub bituminous coal seam, the results estimate the position and the average temperature between (477-727°C) to be obtained with respect to forward gasification. The LFC (liquid from coal)[3] technology is built around a mild pyrolysis or mild gasification process that involves heating the coal under carefully controlled conditions. The Commonwealth Scientific and Industrial Research Organization of (CSIRO) energy technology and the university of NSW(New South Wales, Sydney, Australia)[1], have undertaken to develop a numerical model of the UCG processes aimed for simulating the physico-chemical process together with cavity growth and, ultimately, strata collapse. Serio [6] has been a principal investigator on experimental and/or modeling studies on pyrolysis, gasification, liquifaction, and combustion of coal which include characterization and development of advanced fuel formulation and pyrolysis of material. He developed advanced fuel research widely used FG-DVC model (Functional-Group, Depolymerization, Vaporization, Cross-linking Model) [6]. The system allows the sample to be heated on a pre-programmed temperature profile, at rates 3–100°C min⁻¹, up to a temperature between 20°C and 1100 °C. Isothermal steps with a specified hold time are also possible. The system continuously monitors the time-dependent

evolution of the gases (including specific identification of the individual species such as, CH₄, C₂H₆, C₂H₄, C₂H₂, C₃H₈, benzene, etc...) [7]. The performance of corresponding in-situ pyrolysis project of coal, biomass and waste materials is depending decisively on how the heating of coal and the surrounding rocks can be accomplished through applying suitable heating sources. The suggested heating sources may be point, flat plate, or spherical shaped nuclear bodies of high surface temperatures. They can be placed either in the underground rock layers or directly in coal seams, this study is used for determining the temperature distribution in rock and coal layers to calculate the amount of heated volume of coal which be able to gasify and the heating velocity in coal layer.

2.MODEL DESCRIPTION

Coal is found as a layers between other layers of ground mostly different types of rocks with different thickness, thermal conductivities, densities, specific heats, and different thermal diffusivities, some of physical properties of solids at corresponding temperature such as density, specific heat at constant pressure, thermal conductivity, and thermal diffusivity are given in table (1). The problem of unsteady-state heat conduction in the coal and rock seams is considered. In this model, the heat source is taken as a spherical shaped reactor of radius "R"(m) and constant temperature "T₀"(K) which inserted at different positions. Three cases are studied, in the 1st case the heat source is inserted directly in the coal seams of thickness D_c with Initial Temperature (T_i) not surrounded with rock), Fig.(1.a). The 2nd case

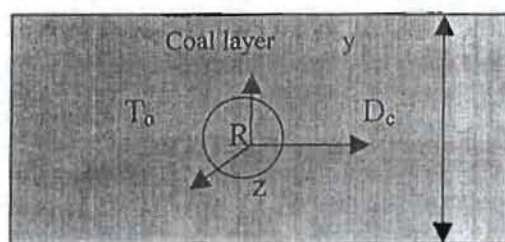
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the heat source is inserted directly in the coal seams between two parallels rock seams of thickness "D_r"(m), Fig.(1.b). While for the third case the heat source is inserted into a rock layer of thickness "D_r"(m)

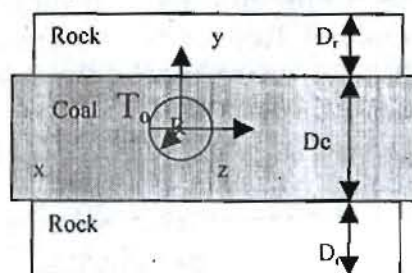
between two parallels coal seams of thickness "D_c"(m), Fig.(1.c). For the problem treatment, the Cartesian coordinate system(x, y, z) is selected, where its position is shown Fig.1

Table (1) Properties of solids from (Adrian Bejan , heat transfer appendix B).

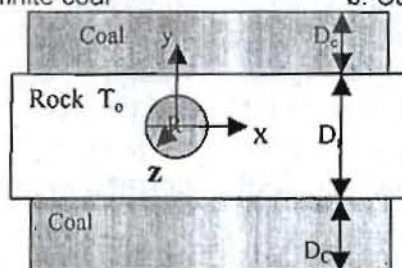
Coal properties Types	Temperature T(°C)	Density ρ (kg/m ³)	Specific heat Cp (KJ/(kg.K))	Thermal conductivity k (W/m.K)	Thermal diffusivity α (cm ² /s)
Coal	20	1200-1500	1.26	0.26	0.0014-0.0017
Anthracite	900	1500		0.2	
Brown coal	900			0.1	
Bituminous in-situ		1300		0.5-0.7	0.003-0.004
Dust	30	730	1.3	0.12	0.0013
Concrete, made with gravel, dry	20	220	0.88	1.28	0.0066
Cinder	20			0.76	
Granite	20	2750	0.89	2.9	0.012
Sandstone	20	2150-2300	0.71	1.6-2.1	0.01-0.013



a. Case 1: HS inserted in infinite coal



b. Case 2: HS inserted in coal layer



c. Case 3: HS inserted in rock layer

Fig.1 The cases of different positions of spherical type heat source (HS) with the selected Cartesian coordinate system

3. THEORETICAL ANALYSIS

The basic differential equation for determining the temperature fields in both rock and coal layers is obtained by coupling the energy equation and Fourier's law of heat conduction. This yields the following differential equation describing the temperature distribution for one dimension, unsteady heat conduction, where the applied heat source causes constant

surface temperature at the inner boundary is described as :

$$\rho c \frac{\partial T}{\partial t} = \frac{\partial}{\partial x} k \frac{\partial T}{\partial x} \tag{1}$$

For constant properties(ρ, c, k), the previous differential equation is reduced to

$$\frac{\partial T}{\partial t} = \alpha \frac{\partial^2 T}{\partial x^2} \tag{1a}$$

where ($\alpha = k / \rho c$) is the thermal diffusivity. The present model concerns with a spherical type heat source inserted according to different position of fig. 1. In this model, both the heat needed for drying the coal and the heat of chemical reaction as well as the variation of physical properties of rocks and coal with temperature will be neglected for the sake of simplicity. Under these conditions, the governing differential equations which describe the temperatures variation of the underground layers with time in the three dimension (x, y, z). may be given as follows: For rock layer

$$\frac{\partial T_r}{\partial t} = \alpha_r \left(\frac{\partial^2 T_r}{\partial x^2} + \frac{\partial^2 T_r}{\partial y^2} + \frac{\partial^2 T_r}{\partial z^2} \right) \quad (2)$$

For coal layer

$$\frac{\partial T_c}{\partial t} = \alpha_c \left(\frac{\partial^2 T_c}{\partial x^2} + \frac{\partial^2 T_c}{\partial y^2} + \frac{\partial^2 T_c}{\partial z^2} \right) \quad (3)$$

The governing differential equations (2&3) are solved by using the discretization method. The initial and boundary conditions associated with equation (2), (3) are:

$$T = T_i \text{ at } t = 0,$$

$$R \leq \sqrt{(x^2 + y^2 + z^2)} \leq \infty$$

$$T = T_i \text{ at } t > 0, \quad x, y, z \rightarrow \infty$$

$$T = T_o \text{ at } t \geq 0, \quad x^2 + y^2 + z^2 = R^2$$

$$T_r = T_c \text{ at } t \geq 0, \quad y = D_i/2, \quad y = -D_i/2$$

Equations (2&3) are linear partial differential equations of second order of parabolic type. Because these differential equations with initial and boundary conditions are relatively complex, they are solved numerically using the finite difference method (control-volume formulation). In case of heterogeneous (an isotropic) layers, equations (2) and (3) may be stated as:

$$\rho c \frac{\partial T}{\partial t} = \frac{d}{dx} \left(\frac{k \partial T}{\partial x} \right) + \frac{\partial}{\partial y} \left(\frac{k \partial T}{\partial y} \right) + \frac{\partial}{\partial z} \left(\frac{k \partial T}{\partial z} \right) \quad (4)$$

The discretization equation which describes the previously mentioned

differential equations can be written as follows:

$$a_p T_p = a_e T_e + a_w T_w + a_n T_n + a_s T_s + a_t T_t + a_b T_b \quad (5)$$

where T_p is the pole temperature, $T_e, T_w, T_n, T_s, T_t, T_b$ are denoting east, west, north, south, top and bottom respectively, while a is a constant. This discretization equation is derived by integrating the differential equations over the control volume shown in Fig.2 and over the time interval from time "t" to time (t+ Δt), as follows:

$$\rho c \int_a^{a+\Delta x} \int_b^{b+\Delta y} \int_c^{c+\Delta z} \left(\frac{\partial T}{\partial t} \right) dx dy dz dt = \int_a^{a+\Delta x} \int_b^{b+\Delta y} \int_c^{c+\Delta z} \left[\frac{\partial}{\partial x} \left(\frac{k_x \partial T}{\partial x} \right) + \frac{\partial}{\partial y} \left(\frac{k_y \partial T}{\partial y} \right) + \frac{\partial}{\partial z} \left(\frac{k_z \partial T}{\partial z} \right) \right] dx dy dz dt \quad (6)$$

where the order of integration is chosen according to the nature of the term. For representation of the term ($\partial T / \partial t$) we shall assume that the grid-point value of (T) prevails throughout the control volume then, the integration of left hand side of equation (6) yields:

$$\rho c \int_a^{a+\Delta x} \int_b^{b+\Delta y} \int_c^{c+\Delta z} \int_t^{t+\Delta t} \frac{\partial T}{\partial t} dt dz dy dx = \rho c \Delta x \Delta y \Delta z (T_{p,t} - T_{p,t}) \quad (7)$$

Similarly, the integration of the right hand side of equation (6) with respect to x, y and z results in

$$\rho c \Delta x \Delta y \Delta z (T_{p,t} - T_{p,t}) = \int_t^{t+\Delta t} \left[\frac{k_x}{\Delta x} (T_e - T_p) \Delta y \Delta z - \frac{k_x}{\Delta x} (T_w - T_p) \Delta y \Delta z + \frac{k_y}{\Delta y} (T_n - T_p) \Delta x \Delta z - \frac{k_y}{\Delta y} (T_s - T_p) \Delta x \Delta z + \frac{k_z}{\Delta z} (T_t - T_p) \Delta x \Delta y - \frac{k_z}{\Delta z} (T_b - T_p) \Delta x \Delta y \right] dt \quad (8)$$

At this point, we need a concept about how $T_p, T_e, T_w, T_n, T_s, T_t$ and T_b vary with time from time "t" to a new time (t+ Δt). Many assumptions are available in the literatures. One of

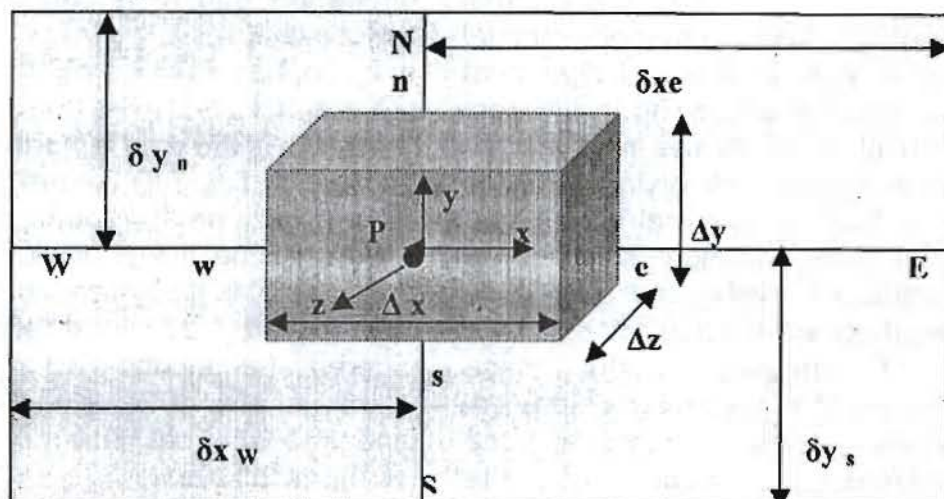


Fig.2 Schematic diagram for the discretization method illustrating nomenclature used in numerical analysis

them can be generalized by the following proposed equations:

$$\int_i^{i+\Delta t} T_p dt = [f T_{pi} + (1-f) T_{pi}] \Delta t \quad (9)$$

Where f is a weighting factor. It has value between 0 and 1.

For fully implicit schemes where ($f = 1$), assume that T_e, T_w, T_n, T_s, T_t vary with time from time " t " to time $(t + \Delta t)$ and the increments of space coordinate are chosen such that: $\Delta x = \Delta y = \Delta z = \Delta L$, by substitution of equ.(9) into equ.(8) then, we have

$$T_{pi} \left[\frac{\rho c \Delta L^3}{\Delta t} + 6k \Delta L \right] = k \Delta L \left(\frac{T_{ei} + T_{wi} + T_{ni} + T_{si} + T_{ti} + T_{bi}}{T_{ni} + T_{si} + T_{ti} + T_{bi}} \right) + \frac{\rho c \Delta L^3}{\Delta t} T_{pi} \quad (10)$$

dividing both sides by ΔL , using

$$\alpha = \frac{k}{\rho c} \quad \text{and let}$$

$$A = (T_{ei} + T_{wi} + T_{ni} + T_{si} + T_{ti} + T_{bi}) \quad (11)$$

We reach to

$$T_{pi} \left[\frac{\Delta L^2}{\Delta t} + 6\alpha \right] = \left[\alpha A + \frac{\Delta L^2}{\Delta t} T_{pi} \right] \quad (12)$$

$$T_{pi} = \frac{\left[\alpha A + \frac{\Delta L^2}{\Delta t} T_{pi} \right]}{\left[\frac{\Delta L^2}{\Delta t} + 6\alpha \right]} \quad (13)$$

For the stability of equation (13), the denominator would be positive

$$\frac{\Delta L^2}{\Delta t} + 6\alpha \geq 0 \quad (14)$$

then $\frac{\Delta L^2}{\Delta t} \leq 6\alpha$ therefore, the time increment can be taken for the sake of simplicity as follows:

$$t = \frac{\Delta L^2}{6\alpha} \quad (15)$$

Then, the following relations hold for the rock and the coal

For the rock

$$\Delta t)_r = \frac{\Delta L^2}{6\alpha_r} \quad (16)$$

For the coal

$$\Delta t)_c = \frac{\Delta L^2}{6\alpha_c} \quad (17)$$

Since $\Delta L_r = \Delta L_c = \Delta L$

Then, The ratio of time increments of both The rock and the coal layers is obtained to be:

$$\frac{\Delta t)_r}{\Delta t)_c} = \frac{\alpha_c}{\alpha_r} \quad (18)$$

by substituting from equation (15) to equation (13), thus

$$T_{p,i} = \frac{A + 6 T_{ps}}{12} \quad (19)$$

Resubstituting from equation (19) into equation (17), the following equation yields :

$$T_{p,i} = \frac{1}{12} \left[\begin{array}{c} T_{e,i} + T_{w,i} + \\ T_{n,i} + T_{s,i} + \\ T_{t,i} + T_{b,i} \end{array} \right] + \frac{1}{2} T_{ps} \quad (20)$$

Often elementary textbooks on heat transfer derive the finite-difference equation via the Taylor-Series method and then demonstrate that the resulting equation is consistent with a heat over a small region surrounding a grid point. We have also see that the control-volume formulation can be regarded as a special version of the method of weighted residuals. The basic idea of the control volume formulation is easy to understand and lends itself to direct physical interpolation. The calculation domain is divided into a number of no overlapping control volumes such that there is one control volume surrounding each grid

Point. The differential equation is integrated over each control volume. Piecewise profiles expressing the variation of (Φ) between the grid points are used to evaluate the required integrals where the (Φ) is the general dependent variable. The result is the discretization equation containing the values of (Φ) for a group of grid points. The discretization equation obtained in this manner expresses the conservation principle for (Φ) for the finite control volume. Just as the differential equation expresses it for an infinitesimal control volume, then applying the discretization equation:

$$a_p T_p = a_e T_e + a_w T_w + a_n T_n + a_s T_s + a_t T_t + a_b T_b$$

When coal or biomass is heated, many reactions including dehydration, cracking, isomerization, dehydrogenation, aromatization, and

condensations take place. the products yields vary, depending on the particular feedstock composition, particle size, heating rate in solids as residence times, and the reactor temperature. For achieving the numerical solution of the discretization equation (control-volume formulation). A developed "Geothermal Processing Schemat Computer Program" is used for calculating the temperature distribution, the gasified volumes of coal and the heating velocity in the coal seams, through solving the differential equations (2&3) associated with the already mentioned initial and boundary associated with the already mentioned initial and boundary conditions. The solution is obtained using spherical heat source of radius R with a surface temperature of (1000°C & 1500°C & 1750°C and 2000°C) which inserted in different positions, Fig.1, in three different types of coal with different composition and properties, table 2. According to heat source position, the temperature of coal layers will increase and the layers are able to gasify when their temperature reaches above (600K).

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Table 2 : different types of coal

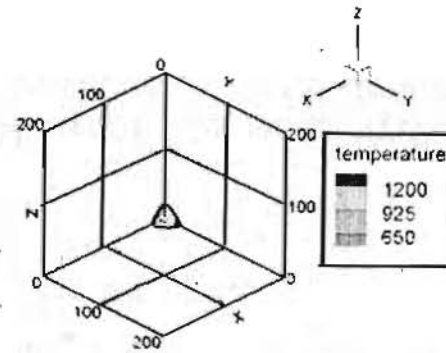
Coal type	type (A) Anthra- cite	type (B) Bitumin- ous	Type (C) Sub- bituminous
Composition			
O	5	3.5	29.5
C	80	60.5	52.5
S	0.7	2	1
H	2.9	3.6	6.29
others	10.5	9	9.8
Thermal conductivity (W/m.K)	0.26	0.35	0.5
Specific heat (KJ/kg.)	1.2	1.26	1.3

3.1 THEORETICAL RESULTS

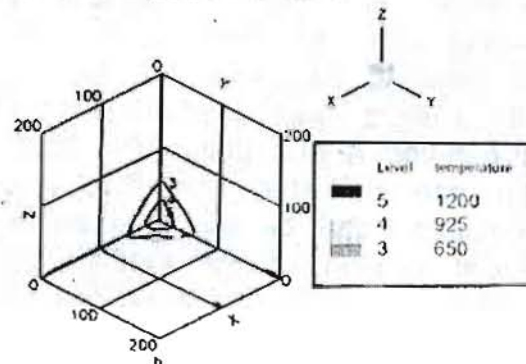
The theoretical solution is done for three considered cases, Fig. 1.

3.1.1 Heat source inserted directly in infinite coal

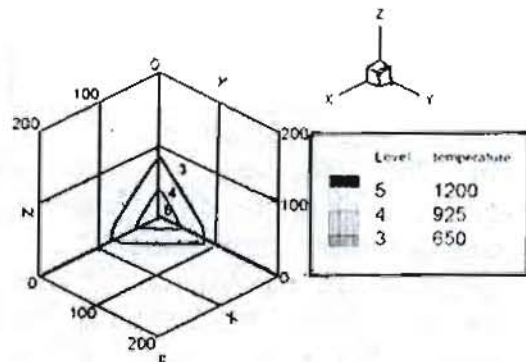
An infinity coal (not surround with rock) of type A (Anthracite) of cubic shape with length 200m is heated by a central spherical heat source with surface temperature 1000°C and 1500°C. The considered case is solved by applying the above developed theoretical method. Figure 3 shows the developing of the coal heated volume and constant temperature contours with time for one day, one week, one month, and two months for heat source with surface temperature of 1000°C. The theoretical solution is performed for the other two types of coal: type B (Bituminous) and type C (Sub bituminous) for the same condition. The obtained results are shown in Figs. 4 and 5.



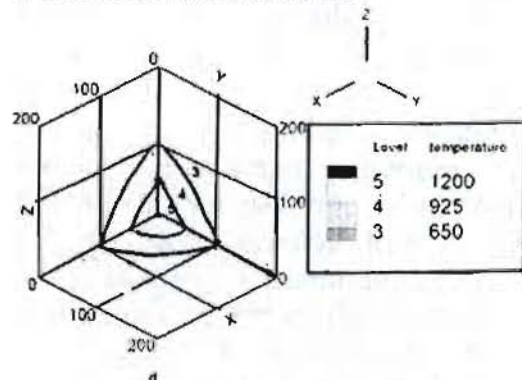
a. One day heated volume



b. One week heated volume

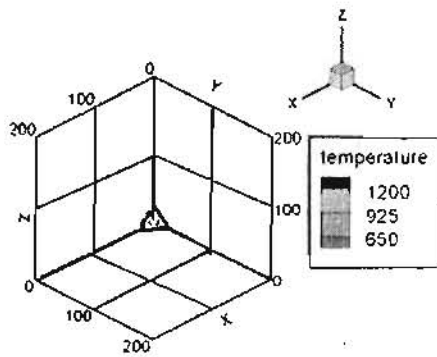


c. One month heated volume

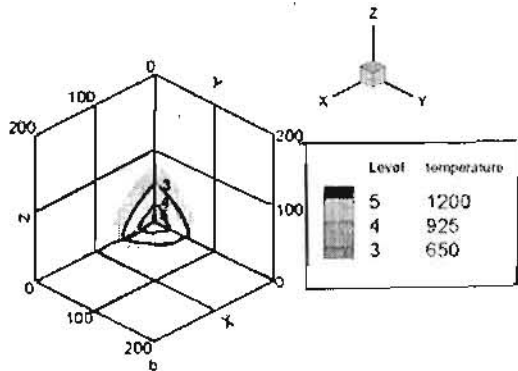


d. Two months heated volume

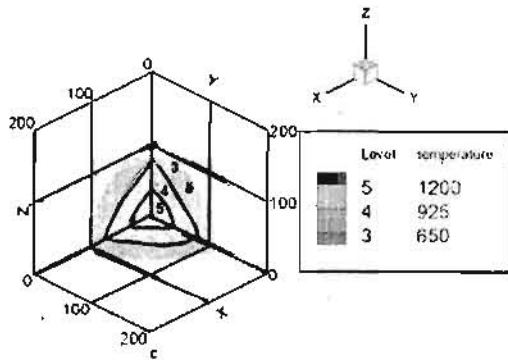
Fig.3 The heated volume and temperature contours variation with time for coal type A at heat source of 1000°C surface temperature



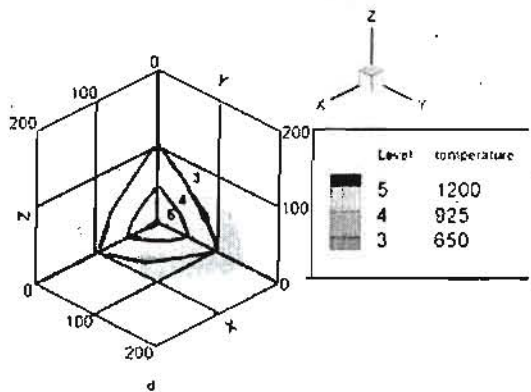
a. One day heated volume



b. One week heated volume

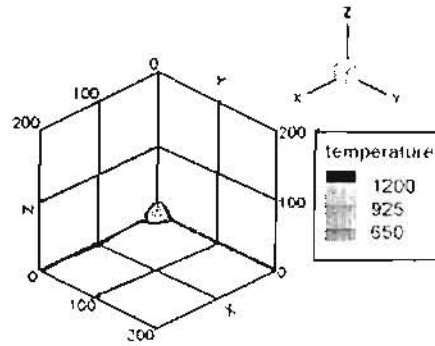


c. One month heated volume

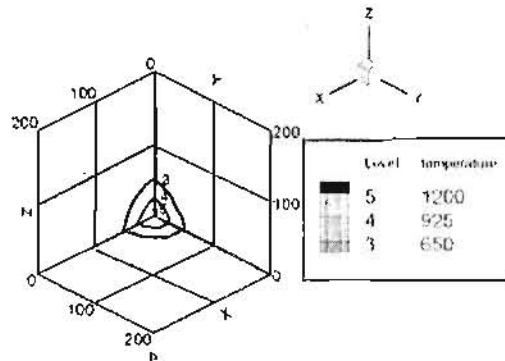


d. Two months heated volume

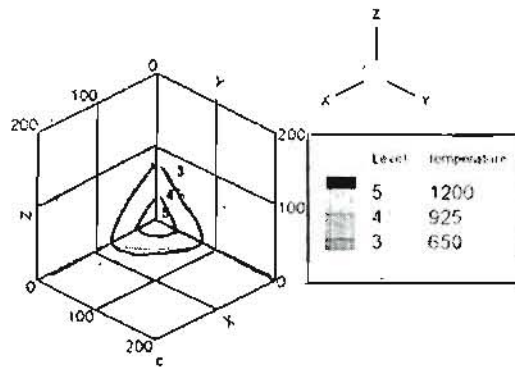
Fig.4 The heated volume and temperature contours variation with time for coal type B at heat source 1000°C



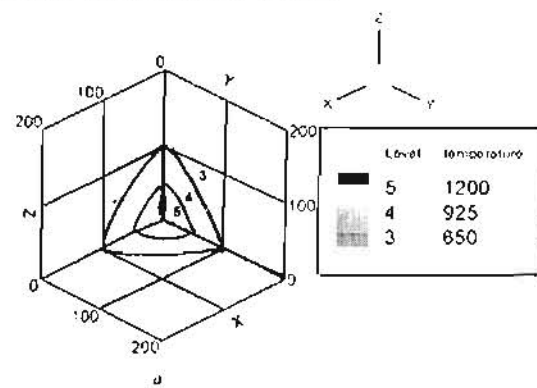
a. One day heated volume



b. One week heated volume



c. One month heated volume



d. Two months heated volume

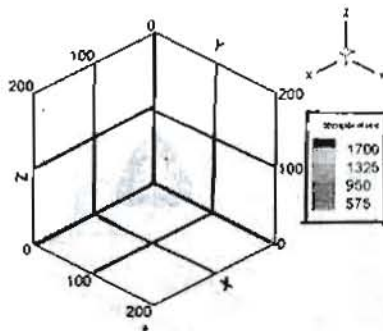
Fig.5 The heated volume and temperature contours variation with time for coal type C at heat source 1000°C

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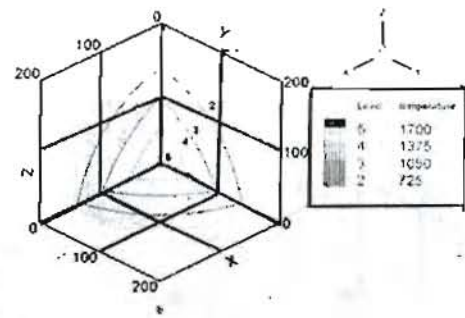
Also the cases with the three coal types are solved theoretically for the heat source with 1500 and 2000°C where the variation of heated volume and temperature contours after one day

and one month period are represented in Fig.6 and Fig. 7 respectively.

Type A

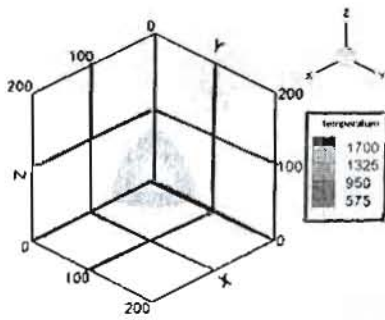


a. one day period

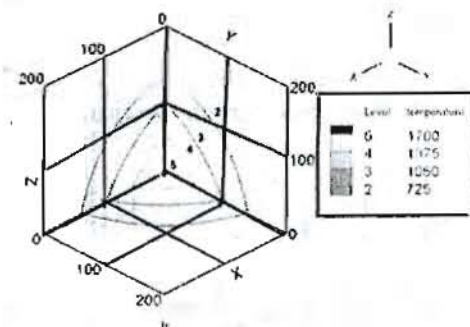


b. one month period

Type B

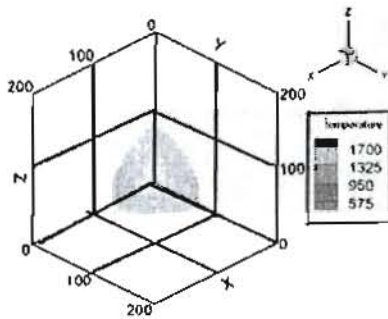


a. one day period

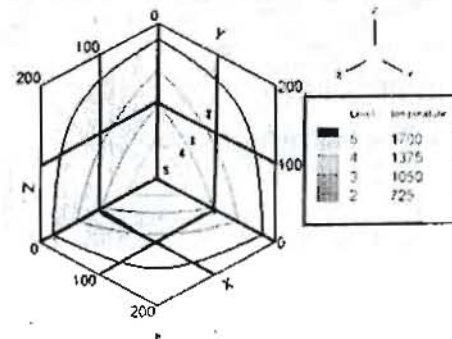


b. one month period

Type C



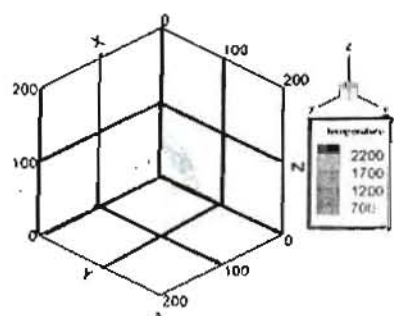
a. one day period



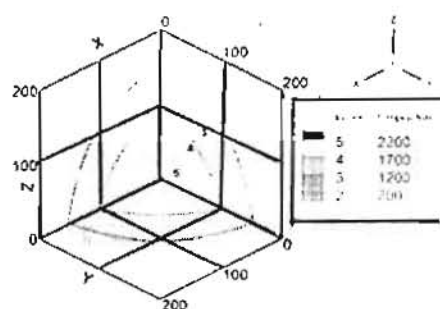
b. one month period

Fig.6. Comparison between the developed heated volume and temperature contours for the three types of coal after one day and one month period for heat source 1500°C

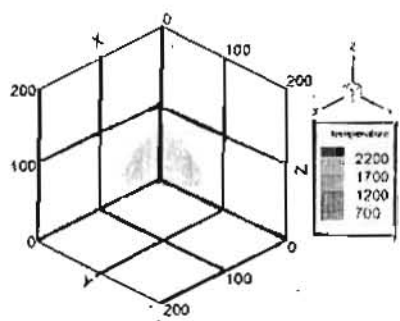
Type A



a. One day period

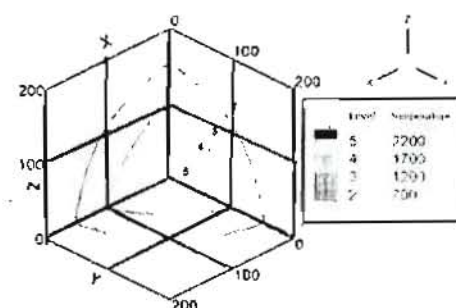


a. One month period



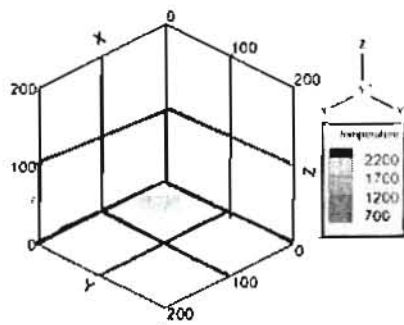
a. One day period

Type B

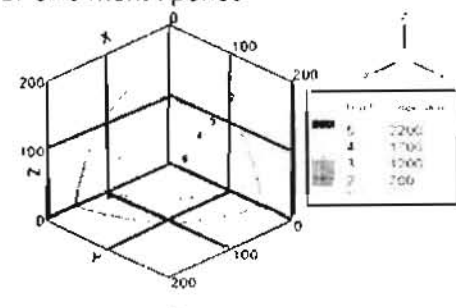


b. One month period

Type C



a. One day period



b. One month period

Fig.7. Comparison between the developed heated volume and temperature contours for the three types of coal after one day and one month period for heat source 2000°C.

Analyzing the calculated results represented in Figs.3-7, it is concluded that the boundaries of heated volume increase with time and with heat source temperature. Also the heated volume of coal increases with thermal conductivity, which depends on coal type. The heated volume increases rapidly at the beginning of trial and after certain period of time, the rate of its increase

starts to decrease till reaching the steady state, at which its boundaries remains unchanged.

3.1.2 Heat source inserted in the coal layer between two rock layers.

Such model is solved as two dimensional problem as the width of coal layer is considered big enough

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compared with coal layer thickness and length. The heat source is inserted in the coal layer at different depth of rock in the range from 150m to 500m, where the rock is granite with properties, density(2750 kg/m³), specific heats (0.89KJ/kg.K), thermal conductivity (2.9 W/m.K) and thermal diffusivity (0.012 cm²/s).

The theoretical solution of this case is applied for coal seam with different thickness range from 40m to 400m and different length range from 250m to 1000m. The heated coal volumes and the temperature contours of two types of coal (type A & C) at heat source temperature 1750°C during different times of two weeks and one month are shown in Fig.8

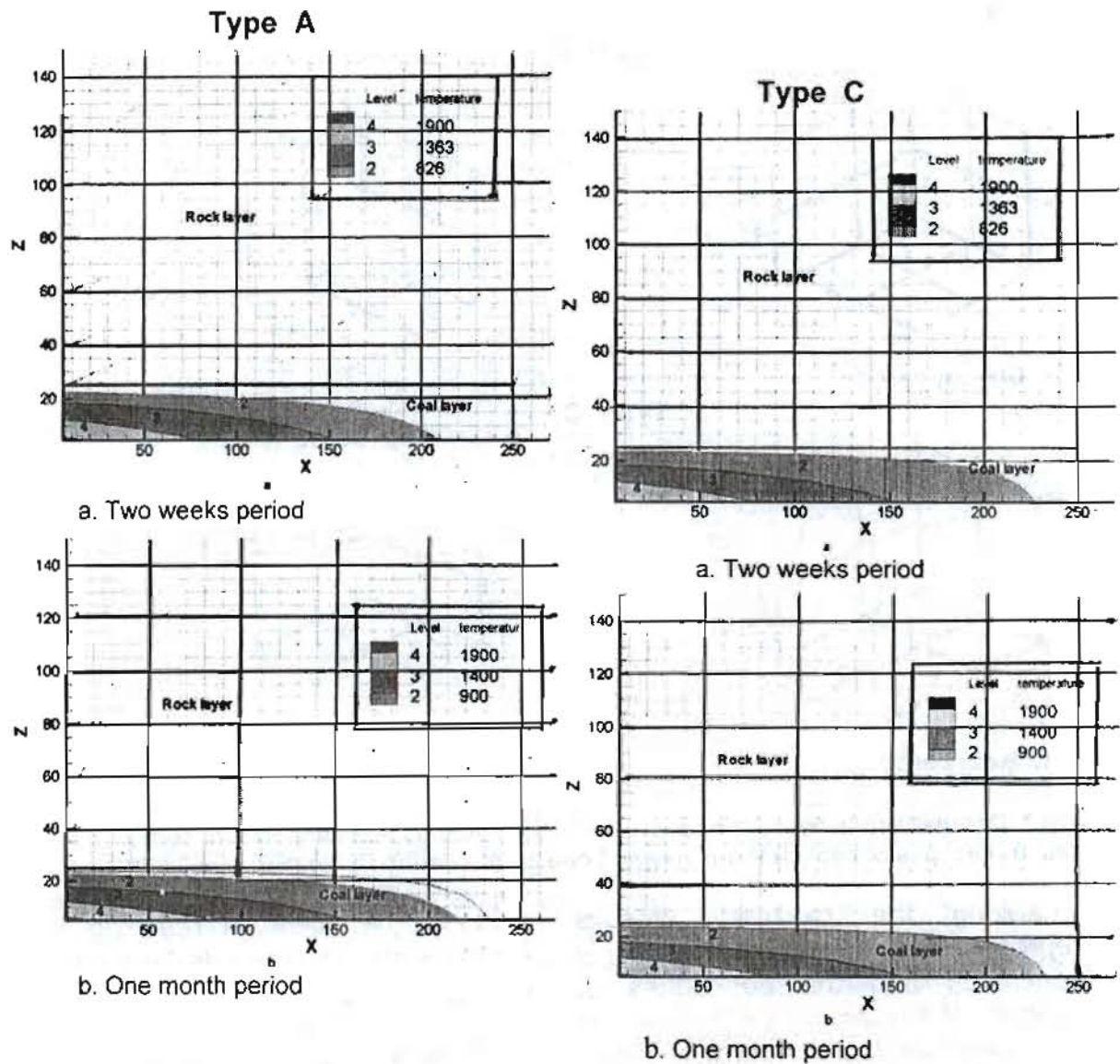


Fig.8 The temperature contours variation with time for coal type(A&C), 40m thickness at depth 150m & heat source 1750°C

Also this model is solved considering coal type C at the same heat source temperature but with depth 500m and

Coal length 1000m, where the obtained results are shown in Fig.9.

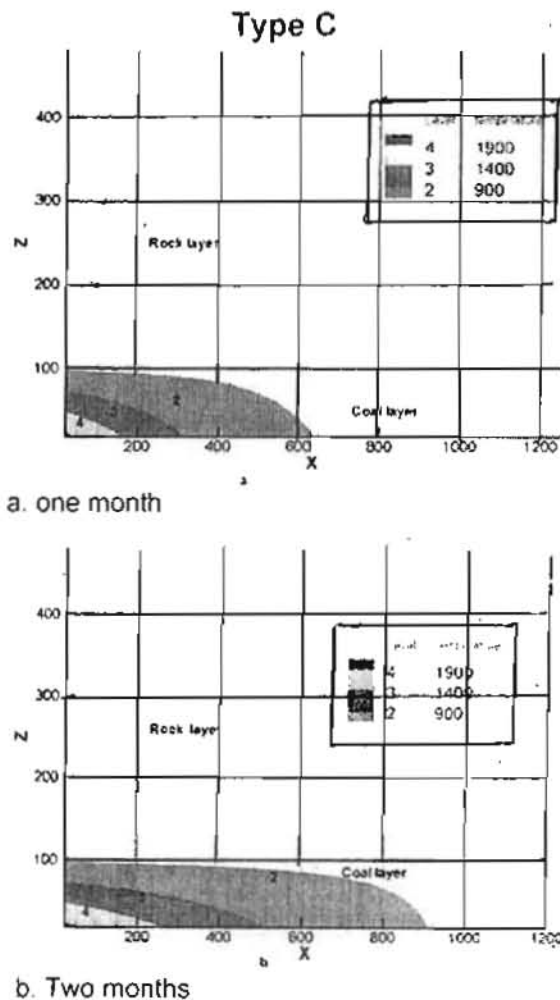


Fig.9 The temperature contours variation with time for coal type C, 200m thickness at depth 500m & heat source 1750°C.

To evaluate the obtained results let us define the heating velocity (C_h) at any position as the rate of change of temperature with time, which given by the following relation :

$$C_h = (dT/dt) = (T_2 - T_1) / \Delta t_{(2-1)}$$

where T_2 is the temperature at new time ($t + \Delta t$),

T_1 is initial temperature,

$\Delta t_{(2-1)}$ is the time interval .

The heating velocity is a criterion by which the change of heated volume is evaluated, such that as C_h approaches zero value, the steady state is reached. The obtained results show that the heating velocity in the

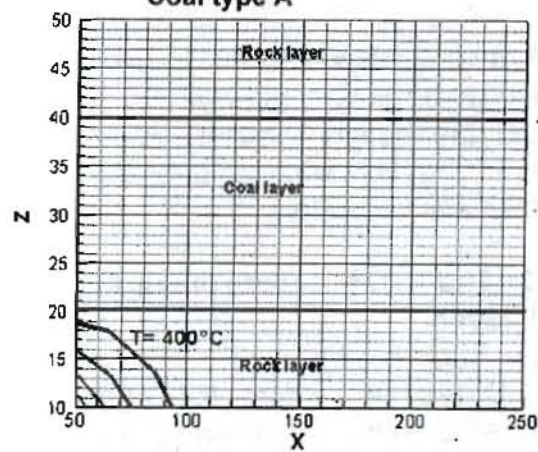
coal layer increases in the beginning of trial with time and after a considered period the heating velocity decreases till reaching The steady state after two months approximately.

3.1.3 The heat source inserted in rock layer

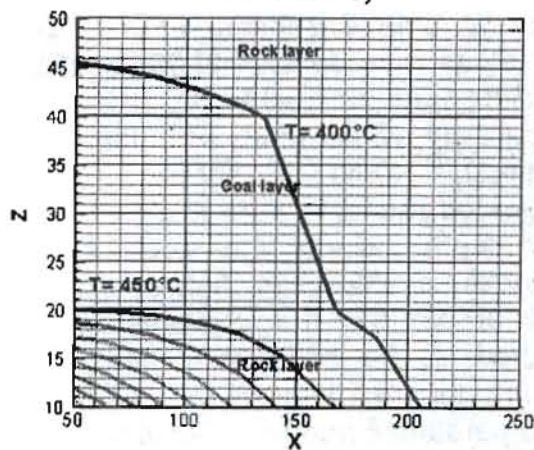
Such model is also considered as 2-D problem where the effect of change of heat source temperature is studied in the range 1000°C to 2000°C. The heat source is inserted in rock of Sandstone with: density(2300 kg/m³), specific heat (0.71kJ/kg.K), thermal conductivity (2.1W/m.k) and thermal diffusivity (0.013 cm²/s). The considered model is 40m thickness at 50m depth, where the coal layer thickness is 20m lies between two layers of rock each 10m thickness. The obtained results is shown in Figs.10, 11 & 12. Figure 10 shows the variation of temperature contours for coal type A with time for one day, one week, and one month, at heat source temperature of 1000°C. While Figs.11 & 12 show the same variation at heat source temperature of 1500°C and 2000°C respectively. The obtained results show that the heated volume of gasified coal layer increase with time and with heat source temperatures where the steady states are reached after one month.

The percentage of gasified volume of coal is the amount of heated volume of coal which be able to gasify from total heated volume which a has temperature $T \geq 400^\circ\text{C}$. The percentage of gasified volume of different coal types (A&B and C) with time at heat source 1000°C, and 2000°C are represented as bar chart in Figs.13 and14.

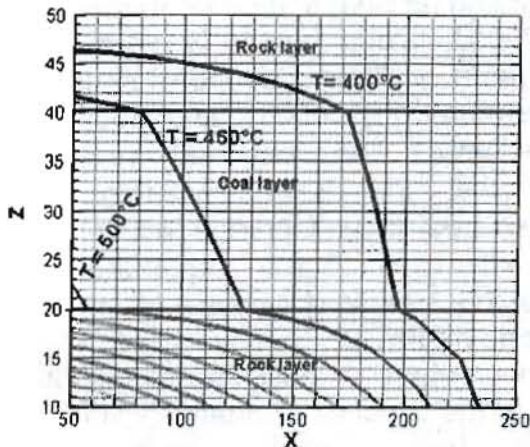
and Eng. Ali Ahmed Ghali
Coal type A



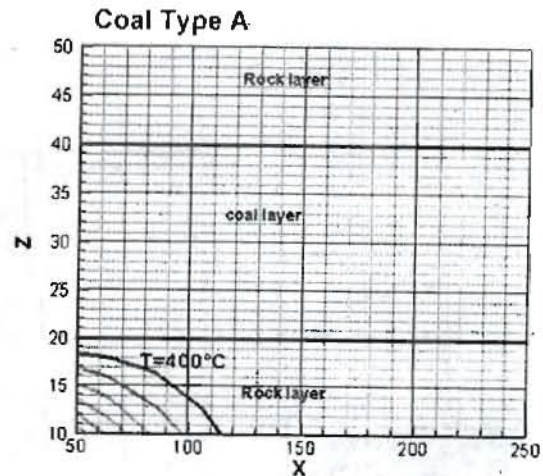
a. One day



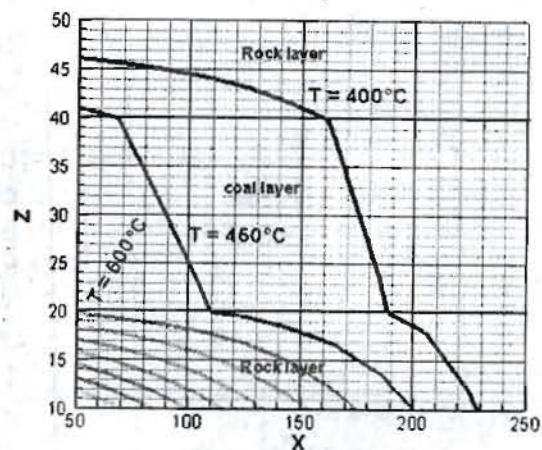
a. One week



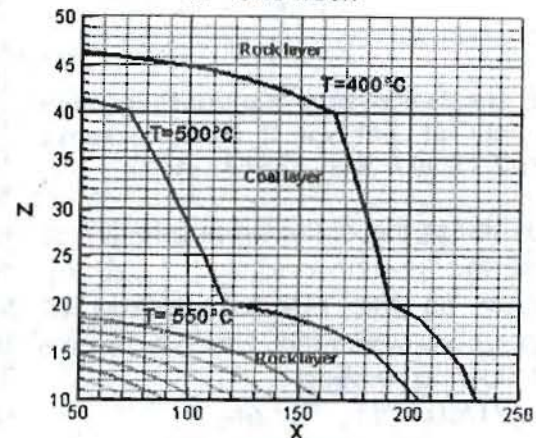
a. One month



a. One day



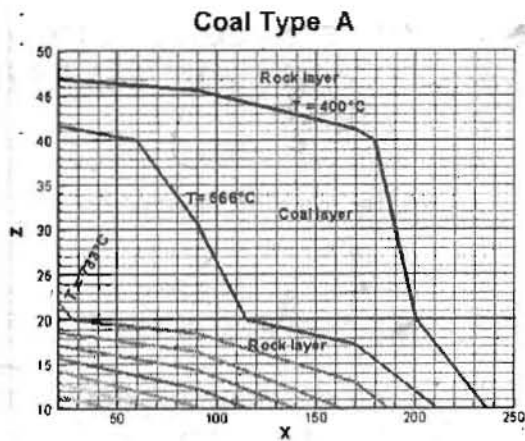
b. One week



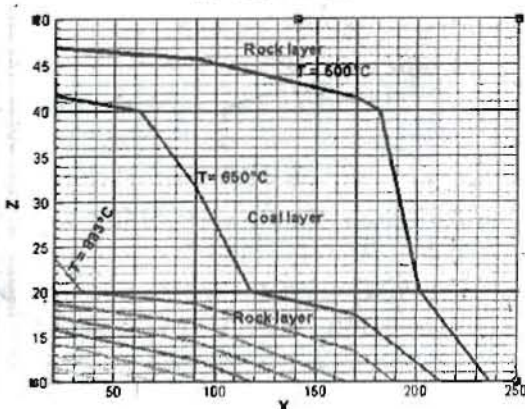
b. One month

Fig.10 Temperature profiles and heated volume of coal at different time periods for heat source with 1000°C inserted in rock layer 250m length.

Fig.11. Temperature profiles and heated volume of coal at different times for heat source with heat source temperature 1500°C inserted in rock layer 250m length.



a. One week



b. One month

Fig.12 Temperature profiles and heated volume of coal at different times, for heat source with temperature 2000°C inserted in rock layer 250 m length.

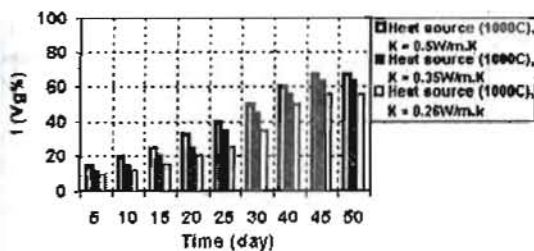


Fig.(13): percentage of gasified volume of different coals at heat source(1000°C) with time

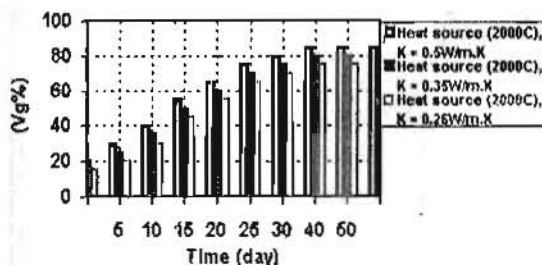


Fig.(14): Percentage of gasified volumes of different coals at heat source(2000°C) with time

The results show that percentage of gasified volume of coal vary with time according to heating velocity in coal layers and the percentage of gasified volume of coal increases in the

beginning of heating pyrolysis trial and reaches constant value at the end of trial depending on the heating velocity and type of coal. While the variation of heating velocity in coal seam with time for different coal types at heat source 1000°C & 1500°C and 2000°C as bar chart are represented in Figs.15 & 16 and 17. These figures indicate that the heating velocity increases with heat source temperature and decreases with time.

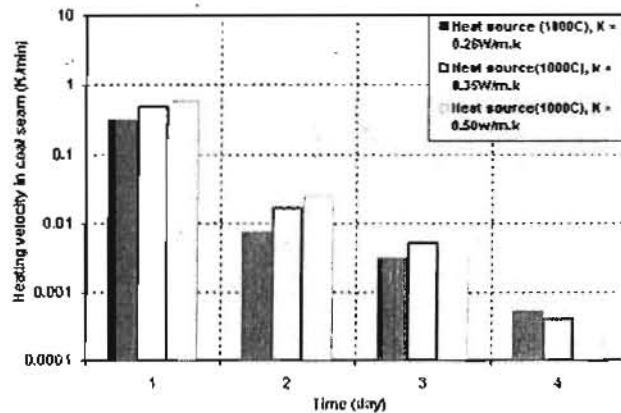


Fig.15 Heating velocity in coal seam with time running and different types of coal at heat source(1000°C)

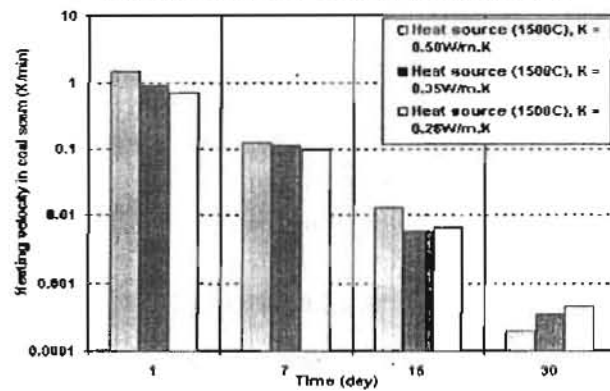


Fig.16 Heating velocity in coal seam with time running and different types of coal at heat source(1500°C)

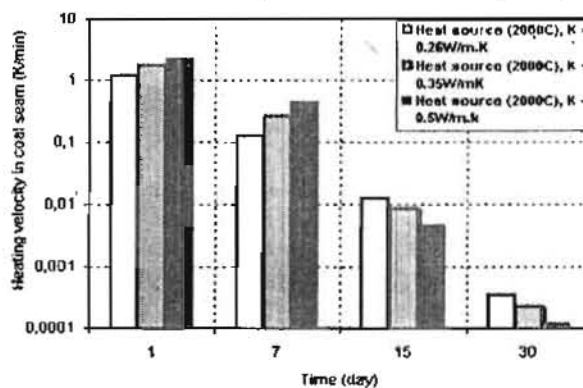


Fig.17 Heating velocity in coal seam with time running and different types of coal at heat source(2000°C)

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4. Efficiency of pyrolysis coal in-situ

The efficiency of coal pyrolysis process is calculated for three cases according to the heating method as follows :

1. Ideal case of homogeneous distribution of heating source.
2. Heating source inserted in infinite coal.
3. Heating source inserted in rock.

Syngas composition results from pyrolysis of underground coal process varies based on many factors, including reactor type, feedstock, and processing conditions (temperature, pressure, type reactant , etc.). The efficiency of coal pyrolysis

(η) is defined as:

$$\eta = \frac{H_{gas} - Q_c}{H_{gas}} = 1 - \frac{Q_c}{H_{gas}} = 1 - \frac{Q_c / m_c}{h_{gas}} \quad (21)$$

and

$$h_{gas} = \frac{H_{gas}}{m_c} \quad (22)$$

Where H_{gas} is the amount of heat content in gasses produced from pyrolysis coal (kJ) which given as:

$$H_{gas} = n_{H_2} V_{H_2} \rho_{NH_2} h_{H_2} + V_{CH_4} \rho_{CH_4} h_{CH_4} + AA \dots (\text{for the gase fuel}) \quad (23)$$

Where m_c is mass of coal pyrolysis.

V_{H_2} is the volume of hydrogen present in gas produced m^3/kg .

V_{CH_4} is the volume of methane present in gas produced m^3/kg .

$$\rho_{NH_2} = \frac{M_{H_2}}{V_N} \quad (24)$$

is the density of hydrogen at normal condition.

$$\rho_{MCH_4} = \frac{M_{CH_4}}{V_N} \quad (25)$$

is the density of methane at normal condition.

h_{gas} is the amount of heat content in gasses produced from pyrolysis coal

per unit mass (kJ/kg), and Q_c is the amount of heat used to heat the coal (kJ) which is given as:

$$Q_c = m_c C_p \Delta T = m_c C_p (T_b - T_i) \quad (26)$$

where C_p is the specific heat of coal (J/kg.K)

T_b is the temperature of heated coal

T_i is the inlet temperature of coal.

4.1 Ideal case of homogeneous distribution of heating source.

For this case table (3) shows the calculated efficiency for different coals at different temperature, where the amount of gases released and its composition at different temperature is taken from [8]. Therefore results is represented in Fig.(18). The calculated efficiencies show that high efficiency at bituminous and lower efficiency at lignite are obtained .

Table 3: pyrolysis efficiency of different coals (lignite, sub bituminous, Bituminous) at different temperatures .

Efficiency η (%)						
T (°C)	400	500	600	700	800	900
Coal type						
Lignite	89.1	80.1	74.53	-	-	59.59
Sub-bituminous coal	73.46	86.7	81.2	75.2	71.2	65.0
Bituminous coal	-	-	91.0	-	86.27	88.7

4.2 Heating source inserted in infinite coal.

Table (4) shows the pyrolysis efficiency of sub bituminous at different temperatures for heating source based on the amount of gases released and its composition [8].

Table 4: pyrolysis efficiency of sub bituminous at different temperatures

Efficiency η (%)						
T(°C)	400	500	600	700	800	900
Sub-bituminous coal	-	65.39	78.62	80.4	77.3	73.1

4.3 Heating source inserted in rock

The heat used for pyrolysis in-situ coal when heating source inserted in rock layer for heating surrounded coal is equal to the heat used for heating rock and the heat used for heating coal. The calculated pyrolysis efficiency of sub bituminous at different temperature for heat source inserted in rock is listed in table 5, based on the amount of gases released and its composition [8].

Table 5: the pyrolysis efficiency of in-situ coal when heating through inserting heat source in rock layer.

Efficiency η (%)						
T(°C)	400	500	600	700	800	900
Sub-bituminous coal	-	34.5	62.8 3	68.1	64.9	60.1

Figure (19) shows the relation between efficiency pyrolysis of in-situ coal at different temperature for different heat source positions. When heating source inserted in rock layer for heating coal, the maximum efficiency is (68.1%) at (700°C) and minimum efficiency is (34.5%) at (500°C). while for heating source inserted directly in coal, the maximum efficiency is (80.4) at (700°C) and minimum efficiency is (65.39%) at temperature(500°C).

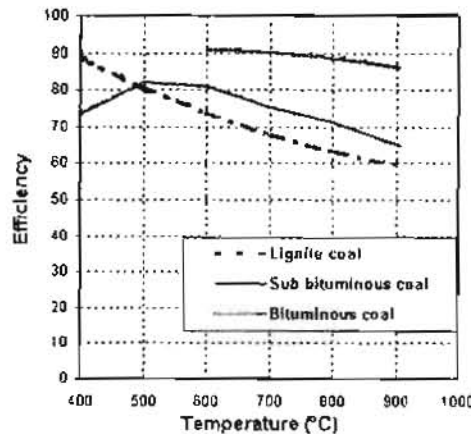


Fig. 18 Pyrolysis efficiency of different coals at different temperature

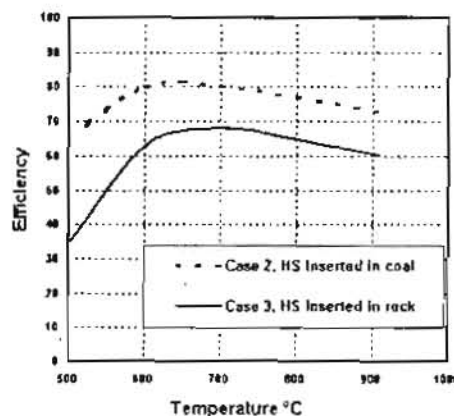


Fig 19 Pyrolysis efficiency of sub bituminous coal with different temperature for different positions

Experimental Investigation

The experimental set up simulates the considered theoretical model. The set-up as shown in fig.20 consists of the following components: an insulated cubic iron tank (1 mm) thickness with dimensions (0.5*0.5*0.5 m) filled with a pure coal of chemical composition (85.5%c, 5%H2, 7%O, 1%S, 1.5%N), type A, about 45 cm of coal height. Heating source is a heater of 3kW power of radius 7.5cm. By using the temperature controller which have four channels, each channel can read and control one temperature. In the experiment the channel one is used for controlling the temperature of heat source installed at position A in the center of cubic, and the other three channels are used for reading only the temperature of positions B, C, and

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D. The used temperature sensors are of type J, which positioned at concentric sphere a-round the heat source at radius 10cm, 15cm, 20cm (position of B, C and D). The pyrolysis produce gases which is collected in a plastic vessel through pipes, valves. A sensitive balance is used to calculate the amount of gas produced and the Chromatography (Agilent 6890 Plus) device in Egyptian Petroleum Research Institute (EPRI) (not shown in figure) is used for analyzing the produced gases. The heating source is a special electrical heater of (3kW) contained in an a spherical shape of iron with radius of 7.5 cm. Using the temperature controller the center of cubic, from heating source temperature is selected 500°C, 800°C and 1000°C. To avoid oxidation of pyrolysis gases and to enhance pyrolysis the space above the coal is evacuated using vacuum pump whereas its absolute pressure reaches 0.8 bar. For each selected temperature source all measured parameters: temperature, gas mass and time are continuously recorded.. Figure 21 shows the measured temperature variation with time at different position of coal seam(heat source A, and the other three position B, C and D) together with theoretically estimated variation at heat source of 1000°C at the same positions. While Fig.22 shows the measured change of heating velocity in 0.45m coal seam with time together with the theoretically calculated change for heat source temperature 1000°C. Both figures show faire agreement between the measured and calculated

results due to the differences properties of the considered physical coal in theoretical and experimental studies. The gas results from pyrolysis process during the time of experimental is the amount of produced gas due pyrolysis at heat source of 1000°C. Also three samples of the produced gas is taken using special traps. These samples are analyzed in Egyptian Petroleum Research Institute (EPRI) using the Gas Chromatography (Agilent 6890 Plus)device. The gas analysis gives: 24.48% CH₄, 6.8% C₂H₆, 8.3% C₃H₈, 13% C₄H₁₀, 11.6% C₅H₁₂, 5.6% C₆H₁₄, 4.5% C₇H₁₆, 7.6% C₈H₁₈ and other hydrocarbons, etc. The results show that CH₄ is essentially a main product of pyrolysis, and this results agree with the results of (A. Williams, M. Pourkashanian, J. M. Jones and N. Skorupska. "Combustion and gasification of coal").[3] The gasification using pyrolysis in this trial starts as the temperature of heated coal layer reaches 400°C, where the released gases flow through pipes are collected in plastic vessel. The mass of the plastic vessel containing released gases are continuously recorded together with time. The gasification rate is calculated using the relation:

$$(m_2 - m_1) / (t_2 - t_1).$$

The experiment is repeated several time, and the gasification rate takes the value (0.078 to 0.082)gm/min. After considerable time (6h), the heating velocity decreases to about zero value, at which the steady state conditions is reached.

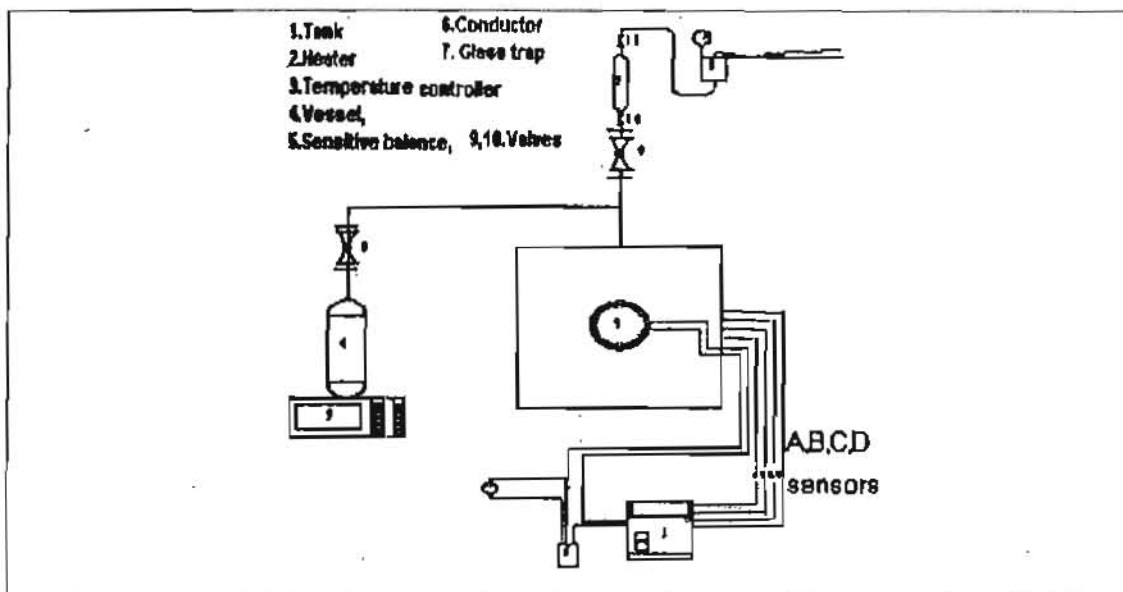


Figure (20): Scheme of the experimental set up.

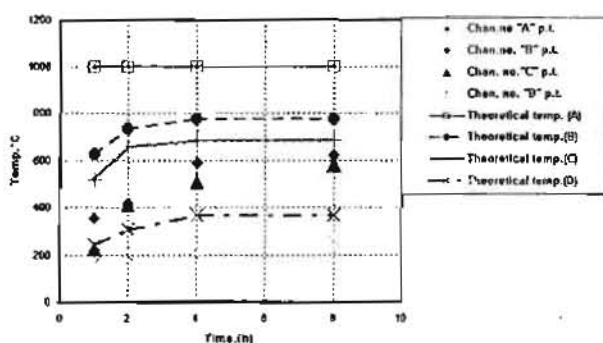


Fig. 21 Temperature distribution in 0.5m coal seam with time at heat source 1000°C, data from programming and experimental set up

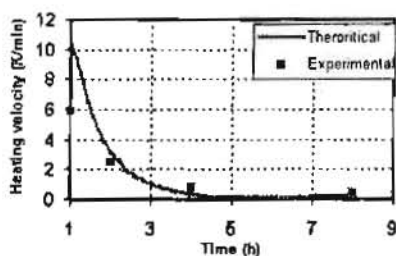


Fig. 22: Heat velocity rate in 0.5m coal layer with time running, data from theoretical programming and experimental set up at heat source 1000°C

Conclusion

The considered theoretical and experimental study of pyrolysis process indicate that:

1.the pyrolysis of underground coal required a complete design and economic analysis of a considerable parameter such as depth, thickness, types and nature of geology. So the important technical requirements and consideration should be taken in

designing a commercial gas production from underground coal gasification in-situ by pyrolysis process such as: a surface temperature of heating source, location and spacing (distance) between the heating sources, heating velocity, production rate of gases, its composition and thermal efficiency, in addition to the developing technology including geological variables,

2.The formulated mathematical model for analyzing the unsteady-state heat conduction during an in-situ pyrolysis of underground coal in order to release gas gives a fair agreement with the experimental results and therefore it can be used for the design and economic analysis of definite pyrolysis process.

3.The temperature distribution in the coal seams depends on the temperature of the heating source, the time running, and the geometry. The heating velocity in coal seams have the mean value between (2 to 10⁻³K/min.) which increases quickly in the beginning of trial, after a considered time it

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decreases and approaches zero at steady state conditions.

4. The operating temperatures of pyrolysis is about 400°C, then thermal decomposition occurs and organic materials are transformed into gases with composition analysis as (CH₄, C₂H₆, C₂H₄, C₂H₂, C₃H₈, H₂, CO₂).

5. The efficiency of in-situ coal pyrolysis depends on the coal compositions where the best temperatures are between (700-800°C) at which the high efficiency of pyrolysis process occurs and high amount of gas produced with high calorific value of heat.

6. This study might be useful in constructing an in situ pyrolysis project for under ground coal.

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