Modeling and controlling for hydraulic excavator's arm

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Abstract—Controlling of excavator's arm is an important and basic task in autonomous research of hydraulic excavator at present, but the behavior of the excavator's arm is dominated by the nonlinear dynamics of the hydraulic actuators. To find a feasible way to control excavator's arm and realize autonomous excavation, firstly, Full kinematic and dynamic models of the excavator arm, regarded as a planar manipulator with three degrees (boom, dipper and bucket) of freedom, were derived, the exponential product formula based on screw theory was used in kinematic model, by which objective angular series of the working mechanism was connected with the desired trajectory of the bucket, and the Lagrange equation was used in dynamic model. secondly, because the experimental excavator's LUDV hydraulic system was not fit for computer control, so the excavator was retrofitted with electrohydraulic proportional valves, associated sensors (three inclinometers), and a computer control system(the motion controller of EPEC), the retrofitted excavator could allow experimental evaluation and refinement of the developed controllers, then ,the full nonlinear math model of electrohydraulic proportional system was achieved; These models included detailed representation of the electrohydraulic actuation system to obtain structural insight into the dynamic behavior of the system; thirdly, According to the highly system dynamic, and parameters uncertainties, external disturbance, the dead region and nonlinear gain coefficient of the proportional direction valve, this paper presents a discontinuous projection based on an adaptive robust controller to approximate the nonlinear gain coefficient of the valve and the nonlinear of the whole system, the error is deal with robust feedback and an adaptive robust controller was designed. Finally, the experiment of the boom motion control is presented to illustrate the feasibility. The thesis covered some phases from modeling to theoretical analysis of controller designs, and to implementation and experimental validation of the new control schemes. These efforts had resulted in new control design methodologies that were applicable to several hydraulic proportional systems when high control performance is required in the presence of severe nonlinearity and uncertainty.

Index Terms—**adaptive robust control, autonomous excavation, dynamic model, hydraulic excavator, kinematic model, motion control**

I. INTRODUCTION

A hydraulic excavator is a multi-functional construction machine. Workers in the construction industry use it for tasks such as excavating, dumping, finishing, lifting work, etc. However, operators who control hydraulic excavators must be trained for many years to do such work quickly and skillfully. A hydraulic excavator has three links: boom, dipper and bucket; and an operator have two arms. Thus, it is not easy for beginners to execute elaborate work that manipulates three links at the same time. Moreover, because the operators have to run work in various dangerous and dirty environments, the number of skillful operators is ever decreasing. For that reason, studying the automation of hydraulic excavators is necessary for improving productivity, efficiency, and safety [1] [2]. The automation of hydraulic excavators has been studied by several researchers in [1]-[6], and so on. Among the fundamental element used as a basis for developing more complicated tasks, the Control of excavator's arm is an important and basic task in autonomous research of hydraulic excavator at present, and this work is difficult from the standpoint of the following problems: parameter variations in mechanical structures, various nonlinearities in hydraulic actuators, and disturbance due to the contact with the ground. For such reasons difficulties exist in controlling a robotic excavator. To solve these problems, several research works have been performed, which may be categorized as either studies based on rules (for instance, in [3] [5]) or model base studies (for example, in [1] [2][4]). This paper presents a discontinuous projection based on an adaptive robust controller to approximate the nonlinear gain coefficient of the valve and the nonlinear of the whole system, the error is deal with robust feedback and an adaptive robust controller is designed. Finally, the experiment of the boom motion control is presented to illustrate the feasibility.

II. OVERVIEW OF EXPERIMENT EXCAVATOR

In this paper, a robotic excavator as a testbed is retrofitted based on the Sunward SWE85 hydraulic excavator, and the hydraulic system retrofitted can be shown in Fig.1. On the base of original hydraulic system, electrohydraulic pilot control has replaced hydraulic pilot control by adding proportional relief valves and substituting hydraulic pilot handle with electrical handle. Using the button on the electrical handle can switch between manual control and computer control.
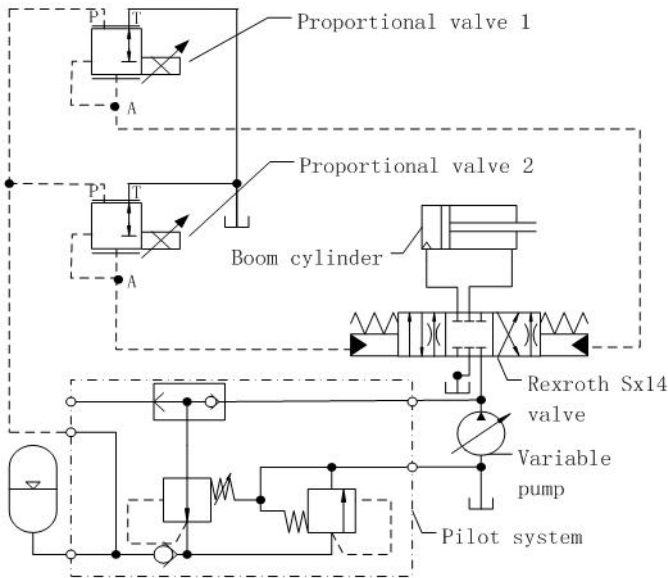
Fig. 1 The retrofitted electrohydraulic system of excavator
(boom cylinder)

The block diagram of hardware designed to control the robotic excavator is seen in Fig.2. The control system consists of two levels; the high-level (PC) controller computes reference angles corresponding to reference trajectories and transfers orders to low-level controller; According to these orders, the low-level controller implements position control by measuring the joints position and pressure of cylinders. That is, controlling the pilot valve to drive the cylinders according to the output current.
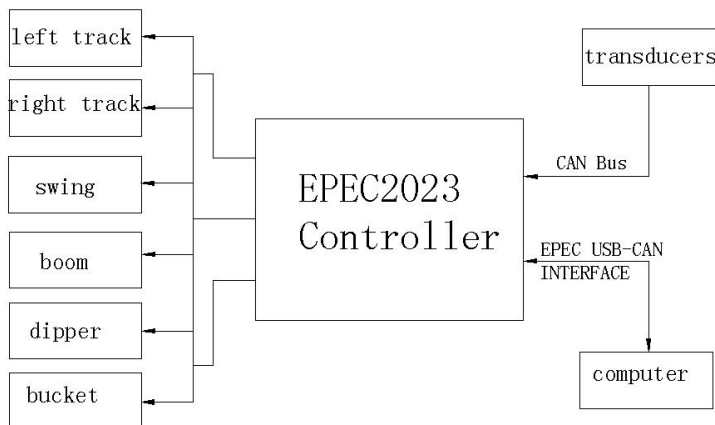

Fig.2 Hardware architecture of the control system

III. MODEL OF ROBOTIC EXCAVATOR'S ARM

The robotic excavator's arm is made up of low level controller, hydraulic actuator system and excavator's manipulator. The manipulator contains boom, dipper and bucket.

A. *The kinematic model of manipulator*

The aim of establishing the kinematic model of manipulator is to transfer bucket tip's reference trajectories to the corresponding and required reference angles sequence of each joint, and to get the motion sequences of hydraulic cylinders.

In the paper, the exponential product formula based on screw theory is used to develop the kinematic model of manipulator. One of the most attractive characteristic of exponential product formula is that it has two coordinate systems, one is base coordinate system S and the other is tool coordinate system T (details in Fig.3).
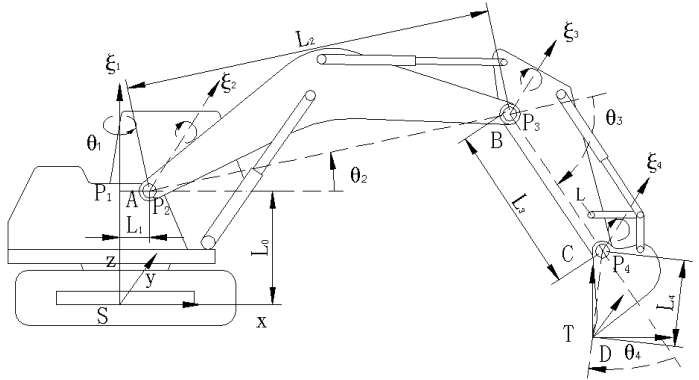

Fig.3 Kinematics model for working mechanism of hydraulic excavator

Given $\omega_1 = \begin{bmatrix} 0 & 0 & 1 \end{bmatrix}^T$　and

$\omega_2 = \omega_3 = \omega_4 = \begin{bmatrix} 0 & 1 & 0 \end{bmatrix}^T$, the motion screws of rotating joints are constructed as follows:

$\xi_1 = \begin{bmatrix} 0 & 0 & 0 & 0 & 0 & 1 \end{bmatrix}^T$, $\xi_2 = \begin{bmatrix} -l_0 & 0 & l_1 & 0 & 1 & 0 \end{bmatrix}^T$,

$\xi_3 = \begin{bmatrix} -l_0 & 0 & l_1+l_2 & 0 & 1 & 0 \end{bmatrix}^T$,

$\xi_4 = \begin{bmatrix} -l_0 & 0 & l_1+l_2+l_3 & 0 & 1 & 0 \end{bmatrix}^T$

Where $\omega_i \in \Re^3$ (i=1, 2, 3) are the unit vectors on the axes direction of motion screws. Then the rigid body transform $g_{st}(\theta)$ is gained relative to the base coordinate system as follows:

$$g_{st}(\theta) = \begin{bmatrix} c_1 c_{234} & -s_1 & c_1 s_{234} & c_1(l_2 c_2 + l_3 c_{23} + l_4 c_{234} + l_1) \\ s_1 c_{234} & c_1 & s_1 s_{234} & s_1(l_2 c_2 + l_3 c_{23} + l_4 c_{234} + l_1) \\ -s_{234} & 0 & c_{234} & l_0 - (l_2 s_2 + l_3 s_{23} + l_4 s_{234} + l_1) \\ 0 & 0 & 0 & 1 \end{bmatrix}$$

$$(1)$$

Where $c_1 = \cos(\theta_1)$　$c_{234} = \cos(\theta_2 + \theta_3 + \theta_4)$,

$s_1 = \sin(\theta_1)$, $s_{234} = \sin(\theta_2 + \theta_3 + \theta_4)$.

$f(t)$ is the reference trajectory of bucket, whose independent variable is t, and the reference position and orientation of bucket can be expressed by a rigid body transform in the base coordinate system. That is:

$$f(t) = \begin{bmatrix} n_x & s_x & a_x & p_x \\ n_y & s_y & a_y & p_y \\ n_z & s_z & a_z & p_z \\ 0 & 0 & 0 & 1 \end{bmatrix} \qquad (2)$$

Where n, s, a is orientation matrix, p is the position vector. The joint angles corresponding to the reference trajectory are obtained by resolving equation (3) (the detail calculating process in [7]).
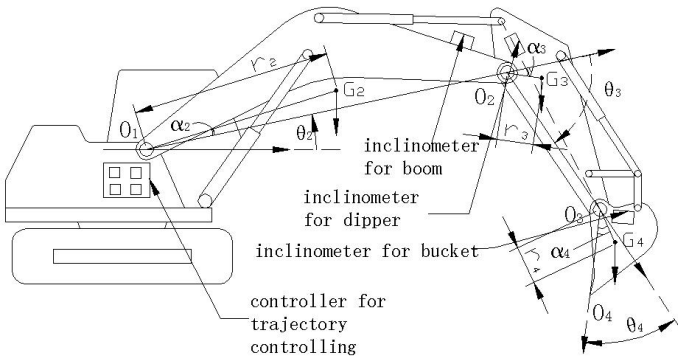
$$f(t) = g_{st}(\theta(t)) \qquad (3)$$

Fig.4 The diagram of manipulator and the position of controller and transducers on the excavator

B. The dynamic model of manipulator

Fig.4 shows a schematic representation of the manipulator of hydraulic excavator with the assigned variables θ_2, θ_3 and θ_4 for the boom, dipper and bucket joint angles, respectively. The cab swing angle (θ_1) is not shown in this figure, as it is not considered in this study. In Fig.4, G_i if the center of gravity for link i expressed in polar coordinates (r_i, α_i) relative to the link i. The link angles with respect to the horizontal plane are θ_2 for the boom, $\theta_{23} = \theta_2 + \theta_3$ for the stick, and for the bucket, $\theta_{234} = \theta_2 + \theta_3 + \theta_4$. The dynamic model can be developed according to Lagrange equation. This paper takes boom as an example, and gives out the dynamic equation [8] - [12] of boom as follows:

$$(D_C + m_L l_e^2)\ddot{\theta}_2 + G_c(\theta_2) + m_L g l_g(\theta_2)$$
$$= \frac{\partial y}{\partial \theta_2}(P_1 A_1 - P_2 A_2) - T(t, \theta_2, \dot{\theta}_2) \qquad (4)$$

$$D_c = I_2 + m_2 r_2^2 + (m_3 + m_4) l_2^2 + m_3 l_2 r_3 \cos(\theta_4 + \alpha_3)$$
$$+ I_3 + m_3 r_3^2 + m_4 l_3^2 + I_4 + m_4 r_4^2 + 2m_4 l_2 l_3 c_3$$
$$+ 2m_4 l_3 r_4 \cos(\theta_4 + \alpha_4) + 2m_4 l_2 r_4 \cos(\theta_{34} + \alpha_4)$$

$$G_c = m_2 g r_2 \cos(\theta_2 + \alpha_2) + (m_3 + m_4) g l_2 \cos\theta_2$$
$$+ m_3 g r_3 \cos(\theta_{23} + \alpha_3) + m_4 g(l_3 \cos(\theta_{23})$$
$$+ r_4 \cos(\theta_{234} + \alpha_4)), l_g = l_2 c_2 + l_3 c_{23} + l_4 c_{234},$$

$$l_e^2 = (l_2 \cos\theta_2 + l_3 \cos\theta_{23} + l_4 \cos\theta_{234})^2$$
$$+ (l_2 \sin\theta_2 + l_3 \sin\theta_{23} + l_4 \sin\theta_{234})^2, i = 2,3,4$$

Where P_1 and P_2 are the pressure of the boom cylinder at the head-side and rod-side respectively. A_1 and A_2 are the areas of pistons on two sides. $T(t, \theta_2, \dot{\theta}_2)$ is the total disturbed moment including moment of load and friction. I_2, I_3, I_4 are moment of inertia of boom, dipper and bucket relative to center of mass of it's own respectively. m_2, m_3, m_4 are mass of boom, dipper, and bucket respectively. m_L is the mass of unknown load in bucket.

C. The model of electro-hydraulic proportional system

1) The hydraulic cylinder equation of continuity

The driving cylinders of the manipulator of hydraulic excavator are control by Rexroth SX14 valve. The driving cylinders of the manipulator of hydraulic excavator are control by Rexroth SX14 valve. The orifice size of directional valve is decided by output pressure of proportional relief valve, which is decided by the output current of controller. Hydraulic system model can be seen in Fig.5.
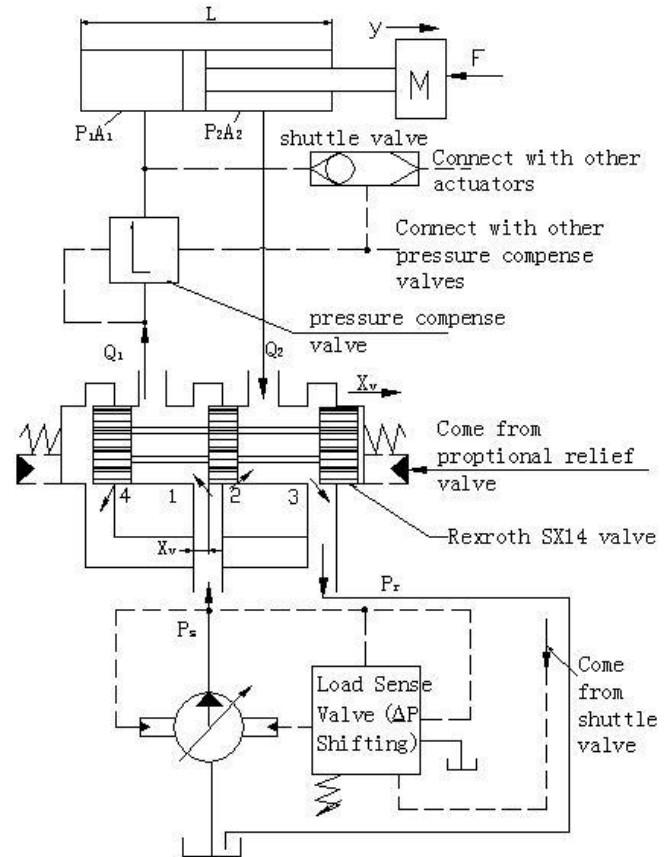

Fig.5 The diagram of LUDV hydraulic system after retrofitting

Following is a summary of the assumption that has been made in developing the model of a hydraulic cylinder:

1) The proportional valve is a symmetrical 3-way and 4-port valve. The dead band of valve is also symmetrical, and the flow in it is turbulent.

2) Possible dynamic behavior of the pressure in the transmission lines between valve and actuator is assumed to be negligible.

3) Pressure is equal everywhere in one volume of hydraulic cylinder, and the temperature and the bulk modulus are constants.

4) The leakage of flows is laminar.

The continuous equation [13] of compressible oil is given as follows:

$$\sum Q_{in} - \sum Q_{out} = \frac{dV}{dt} + \frac{V}{\beta} \bullet \frac{dP}{dt} \qquad (5)$$

Here V is the initial volume of liquid subjected to compression. dV and dP are the changes in pressure and volume respectively. $\sum Q_{in}$ is the input flows of liquid, and $\sum Q_{out}$ is the output flows of liquid. β is the bulk modulus.

Considering the internal and external leakage of cylinder, the equation (6) and (7) are can be developed from equation (5) as follows:

$$Q_1 - C_{ic}(P_1 - P_2) - C_{ec}P_1 = \frac{dV_1}{dt} + \frac{V_1}{\beta_e} \bullet \frac{dP_1}{dt} \qquad (6)$$

$$C_{ic}(P_1 - P_2) - Q_2 - C_{ec}P_2 = \frac{dV_2}{dt} + \frac{V_2}{\beta_e} \bullet \frac{dP_2}{dt} \qquad (7)$$

Where Q_1 and Q_2 are the supplied and return flow to the cylinder respectively. C_{ic} is the internal leakage coefficient. C_{ec} is the external leakage coefficient. β_e is the effective bulk modulus(including liquid, the air in oil and so on). V_1, V_2 are the volume of fluid flow to and from the hydraulic cylinder. V_1, V_2 can be got as follows:

$$V_1 = V_{01} + A_1 y, \qquad V_2 = V_{02} - A_2 y \qquad (8)$$

Where V_{01} is the initial volume of cylinder side which the fluid flows into. V_{02} is the initial volume of cylinder side from which the fluid flows out. y is the displacement of piston. So the equation (9) can be given as:

$$\frac{dV_1}{dt} = A_1 \frac{dy}{dt} \qquad -\frac{dV_2}{dt} = A_2 \frac{dy}{dt} \qquad (9)$$

The external leakage is the leakage between the piston rod and external seals. Because external leakage on the construction machine is forbidden, and external leakage is disappear with the development of sealing technology. The influence of external leakage is neglected in this paper. y can be expressed as a function of θ_2. Then equation (6) and (7) can be rebuilt as:

$$\dot{P}_1 = \frac{\beta_e}{V_1}[Q_1 - C_{ic}(P_1 - P_2) - A_1 \frac{\partial y}{\partial \theta_2}\dot{\theta}_2] \qquad (10)$$

$$\dot{P}_2 = \frac{\beta_e}{V_2}[C_{ic}(P_1 - P_2) - Q_2 + A_2 \frac{\partial y}{\partial \theta_2}\dot{\theta}_2] \qquad (11)$$

Where C_{ic} is assumed to be a constant to simplify the system model.

2) The flow equation of electro-hydraulic proportional valve

In this paper, Load Independent Flow Distribution (abbreviated to LUDV) hydraulic system is adopted on the experimental robotic excavator. According to the theory of LUDV system [14], the flow equation can be gotten as follows:

$$Q_1 = C_d W X_v \sqrt{\frac{2}{\rho}\Delta P_1} = \begin{cases} C_d W K_I I(t)\sqrt{\frac{2}{\rho}\Delta P}, I(t) \geq 0 \\ -C_d W K_I I(t)\sqrt{\frac{2}{\rho}(P_1 - P_r)}, I(t) < 0 \end{cases}$$
(12)

$$Q_2 = C_d W X_v \sqrt{\frac{2}{\rho}\Delta P_2} = \begin{cases} -C_d W K_I I(t)\sqrt{\frac{2}{\rho}(P_2 - P_r)}, I(t) \geq 0 \\ C_d W K_I I(t)\sqrt{\frac{2}{\rho}\Delta P}, I(t) < 0 \end{cases}$$
(13)

Where P_r is the pressure of return oil. ΔP is the pressure of spring of load sense valve. C_d is the flow gain coefficient. W is the area gradient of orifice. X_v is the displacement of spool. ρ is the density of oil. K_I is the current gain of proportional valve. $I(t)$ is the control current of proportional valve.

In this paper, the proportional direction control valve has a quite large dead band, which is due to the close centered valve configuration. C_d is proved to be nonlinear by experiment [15]. The net actual valve opening x_{nav} can be defined as

$$x_{nav} = \begin{cases} x_v - x_d & , x_v > x_d \\ 0 & , -x_d \leq x_v \leq x_d \\ x_v + x_d & , x_v < -x_d \end{cases} \qquad (14)$$

Where x_d is the dead band of valve.

The actual experimental test results reveal the nonlinear flow gain in the Fig.6, and the flow gain coefficient of valve can be approximated by two lines to simplify the model. To be controlled conveniently, Q (the flow of valve) can be expressed

as equation (15) and (16).

$$Q(x_v, \Delta P) = Q_M(x_v, \Delta P) + \tilde{Q}(x_v, \Delta P) \qquad (15)$$

$$Q_{1M} = C_{d1} W f_1(\Delta P_1) x_{nav}$$
$$Q_{2M} = C_{d2} W f_2(\Delta P_2) x_{nav} \qquad (16)$$

where Q_M represent certain simplified flow mapping function that will be used in the controller design. $\tilde{Q}$ is the model error of the flow projection, and will be treated by robust feed back. Generally speaking, C_{d1} and C_{d2} are constant for one working state.
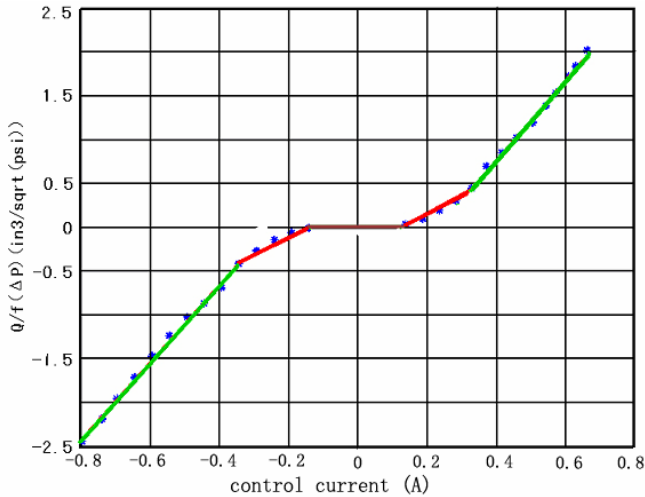

Fig.6 nonlinear characteristic flow gain of the proportional valve

IV.   THE ADAPTIVE ROBUST CONTROLLER

According to the model of excavator, the main difficulties to control the robotic excavator are that:(1)the dynamic of system is strongly nonlinear; (2) parameters of system are uncertain, for example, the uncertainty of m_L, β_e (valid bulk modulus) and T_n (the total disturb of system) which is brought by external disturb and un-modeled friction.

In this paper, the nonlinearity and uncertainty of parameters are treated by a discontinuous projection based on an adaptive robust controller [9]. In Fig.6, the line is used to compensate for the dead band of vale, which is the symmetrical line of plus curve based on the axis that is through the coordinate origin and make an angle of 45 8with the x axis. The flow projection error is compensated by robust feed back. m_L, β_e and T_n are mainly considered as

$$q_1 = \frac{1}{1 + (l_e / D_c) m_L}, q_2 = \frac{T_n}{D_c + m_L l_e^2} \qquad q_3 = \beta_e \text{, and}$$

$q = [q_1, q_2, q_3]^T$ is defined as unknown parameter set. Then equation (4), (10) and (11) can be transform to equation (17).

$$\ddot{\theta}_2 = \frac{q_1}{D_c} [\frac{\partial y}{\partial \theta_2} (P_1 A_1 - P_2 A_2) - G_c(\theta_2)]$$

$$+ \frac{q_1}{l_e^2} g l_g - \frac{1}{l_e^2} g l_g + q_2 + \tilde{T}(t, \theta_2, \dot{\theta}_2), \qquad (17)$$

$$\dot{P}_1 = \frac{q_3}{V_1} (-C_{ic}(P_1 - P_2) - A_1 \frac{\partial y}{\partial \theta_2} \dot{\theta}_2 + Q_{1M} + \tilde{Q}_1),$$

$$\dot{P}_2 = \frac{q_3}{V_2} (A_2 \frac{\partial y}{\partial \theta_2} \dot{\theta}_2 - Q_{1M} - \tilde{Q}_2 + C_{ic}(P_1 - P_2))$$

Where, $\tilde{T} = \dfrac{T(t, \theta_2, \dot{\theta}_2) - T_n}{D_c + m_L l_e^2} \qquad q_{min} \le q \le q_{max}$

$|\tilde{T}| \le \delta_T, |\tilde{Q}_1| \le \delta_{Q1}, |\tilde{Q}_2| \le \delta_{Q2}$, and q_{min}, q_{max} δ_T

δ_{Q1} and δ_{Q2} are known number. It is assumed that $\hat{q}$ and $\tilde{q}$ are the estimated value and estimated error of q respectively. Equation (18) is the adaptation law structure needed.

$$\dot{\hat{q}} = \text{Pr}\, oj_{\hat{\theta}}(\Gamma \tau) \qquad (18)$$

Where Γ is the diagonal matrix. τ the adaptive function. $\text{Pr}\, oj_{\hat{\theta}}(\bullet)$ is the discontinuous projection [16].

According to the above work of this paper, the following task must be undertaken.

Step 1

Defining $P_L = P_1 A_1 - P_2 A_2$, P_L can be treated as the input of equation   17   . The goal of this step is to get a virtual control law P_{Ld} for P_L so that output error $z_1 = \theta_2 - \theta_{2d}$ can converge to zero to get high trajectory tracking accuracy.

Assuming that $\dot{\theta}_{2r} = \dot{\theta}_{2d} - k_1 z_1 \qquad \ddot{\theta}_{2r} = \ddot{\theta}_{2d} - k_1 \dot{z}_1$

$$z_2 = \dot{\theta}_2 - \dot{\theta}_{2r}, \phi_2 = \left[\frac{1}{D_c} (\frac{\partial y}{\partial \theta_2} P_{Lda} - G_c) + g \frac{l_g}{l_e^2}, 1, 0 \right]^T$$

and according to function of $Lyapunov$, the following equation can be gained.

$$P_{Ld} = P_{Lda} + P_{Lds}, \quad \tau_2 = \phi_2 z_2,$$

$$P_{Lda} = \frac{\partial \theta_2}{\partial y} [G_c(\theta_2) + \frac{D_c}{q_1} (-g \frac{\hat{q}_1}{l_e^2} l_g + g \frac{1}{l_e^2} l_g - \hat{q}_2 + \ddot{\theta}_{2r})]$$

$$P_{Lds} = P_{Lds1} + P_{Lds2}$$

$$P_{Lds1} = -\frac{D_c}{q_{1min}} \frac{\partial \theta_2}{\partial y} (k_{2s1} + k_2) z_2, k_2 > 0 \qquad (19)$$

Where k_{2s1} is a control gain to be synthesized later, P_{Lds2} is any function satisfying the following conditions

Condition1 $z_2 [\frac{1}{D_c} q_1 \frac{\partial y}{\partial \theta_2} P_{Lds2} - \tilde{q}^T \phi_2 + \tilde{T}] \le \varepsilon_2$

$\qquad (20)$

Condition2 $z_2 \frac{\partial y}{\partial \theta_2} P_{Lds2}, \varepsilon_2 > 0$

Step 2

$z_3 = P_L - P_{Ld}$ is assumed to be an error of input then the actual controlling principle will be developed to make z_3 approximate to zero. According to equation (17)   (19), the equation (21) can be given as follows:

$$\overset{\circ}{z}_3 = \overset{\circ}{P}_L - \overset{\circ}{P}_{Ld} = q_3[-(\frac{A_1^2}{V_1} + \frac{A_2^2}{V_2})\frac{\partial y}{\partial \theta_2}\overset{\circ}{\theta}_2 \qquad (21)$$

$$+ (\frac{A_1}{V_1}Q_{1M} + \frac{A_2}{V_2}Q_{2M}) + (\frac{A_1}{V_1}\tilde{Q}_1 + \frac{A_2}{V_2}\tilde{Q}_2)] - \overset{\circ}{P}_{Ldc} - \overset{\circ}{P}_{Ldu}$$

Where $\overset{\circ}{P}_{Ldc} = \frac{\partial P_{Ld}}{\partial \theta_2}\overset{\circ}{\theta}_2 + \frac{\partial P_{Ld}}{\partial \overset{\circ}{\theta}_2}\overset{\circ\circ}{\theta}_2 + \frac{\partial P_{Ld}}{\partial t}$ is the calculable

part of $\overset{\circ}{P}_{Ld}$, and $\overset{\circ}{P}_{Ldu} = \frac{\partial P_{Ld}}{\partial \overset{\circ}{\theta}_2}[-\frac{1}{D_c}(\frac{\partial y}{\partial \theta_2}P_L - G_c(\theta_2))\tilde{q}_1$

$-\frac{1}{l_e^2}gl_g\tilde{q}_1 - \tilde{q}_2 + \tilde{T}] + \frac{\partial P_{Ld}}{\partial q}\overset{\circ}{q}$ is the error part of

$\overset{\circ}{P}_{Ld}$ .this error part is treated by robust feedback method.

$\frac{A_1}{V_1}Q_{1M} + \frac{A_2}{V_2}Q_{2M} = Q_L$ is considered as a valid input in

equation (21). In this process, Q_{Ld} which is the valid input of Q_L must be developed to make P_L track P_{Ld} which is developed in step (1). So Q_{Ld} (controlling function) and τ_3 (adaptive function) can be given respectively as follows:

$$Q_{Ld}(\theta_2, \overset{\circ}{\theta}_2, P_1, P_2, \hat{q}, t) = Q_{Lda} + Q_{Lds} \qquad \tau_3 = \tau_2 + \phi_3 z_3$$

$$Q_{Lda} = -\frac{1}{q_3}Q_{Lde} \qquad Q_{Lds} = Q_{Lds1} + Q_{Lds2} \qquad (22)$$

$$Q_{Lds1} = -\frac{1}{q_{3\min}}(k_3 + k_{3s1})z_3, k_3 > 0$$

Where

$$Q_{Lde} = \frac{1}{D_c}\frac{\partial y}{\partial \theta_2}z_2\hat{q}_1 - \hat{q}_3(\frac{A_1^2}{V_1} + \frac{A_2^2}{V_2})\frac{\partial y}{\partial \theta_2}\overset{\circ}{\theta}_2 - \overset{\circ}{P}_{Ldc},$$

$$\phi_3 = [\frac{1}{D_c}\frac{\partial y}{\partial \theta_2}z_2 - \frac{\partial P_{Ld}}{\partial \dot{\theta}_2}[\frac{1}{D_c}(\frac{\partial y}{\partial \theta_2}P_L - G_c(\theta_2))$$

$$+ \frac{1}{l_e^2}gl_g], \frac{-\partial P_{Ld}}{\partial \overset{\circ}{\theta}_2}, -(\frac{A_1^2}{V_1} + \frac{A_2^2}{V_2})\frac{\partial y}{\partial \theta_2}\overset{\circ}{\theta}_2 + Q_{Lda}]^T$$

Where k_{3s1} is a control gain to be synthesized later, P_{Lds2} is any function satisfying the following conditions:

Condition 1 $z_3[q_3 P_{Lds2} + q_3\tilde{Q}_L - \tilde{q}^T\phi_3 - \frac{\partial P_{Ld}}{\partial \theta_2}\tilde{T}] \leq \varepsilon_3$

Condition 2 $z_3 Q_{Lds2}, \varepsilon_3 > 0$

(23)

After Q_{Ld} being gained, the valid displacement of spool can

be computed according to equation (12), (13) and (16) as follows:

$$x_{nav} = \frac{Q_{Ld}(\theta_2, \overset{\circ}{\theta}_2, P_1, P_2, \hat{q}, t)}{\frac{A_1}{V_1}C_{d1}f_1(\Delta P_1) + \frac{A_2}{V_2}C_{d2}f_2(\Delta P_2)} \qquad (24)$$

x_v is assumed to be proportionable to $I(t)$, that is ,

$x_v = KI(t)$ then

$$I(t) = Kx_v = \begin{cases} K(x_{nav} + x_d) & , x_{nav} > 0 \\ 0 & , x_{nav} = 0 \\ K(x_{nav} - x_d) & , x_{nav} < 0 \end{cases} \qquad (25)$$

Assuming $\tau = \tau_3$ in equation (18), k_{2s1} and k_{3s1} are selected to make

$$k_{2s1} \geq \frac{1}{2}\|\Gamma\phi_2\|^2, k_{3s1} \geq \|\partial P_{Ld}/\partial \hat{q}_2\|^2 + \frac{1}{2}\|\Gamma\phi_2\|^2 \qquad (26)$$

As a result, high accuracy can be achieved by using equation (25) as a controlling principle.

V. EXPERIMENT AND CONCLUSION

The motion controlling experiment of boom is done on the robotic excavator, which is based on the Sunward SWE85 excavator. It is assumed that the reference angle is a sine curve, whose amplitude is 40 degrees, and whose center line is -2 degree. The experimental result is showed in Fig.7. The result exhibits good tracking performance for boom cylinder under the controller developed in this paper. The peak error is less than 4 degrees.
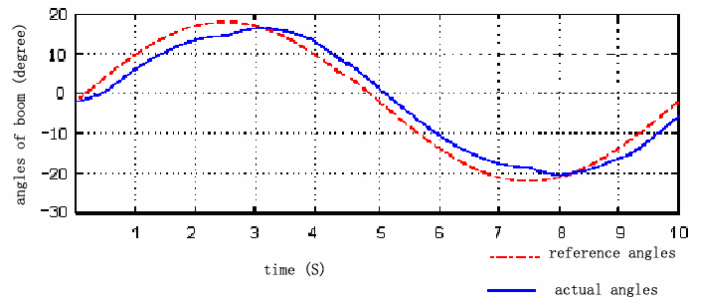

Fig.7 experiment of the boom's angle control

The motion control of hydraulic excavator is the base to achieve automatic excavation. The motion control principle is developed in this paper, under which good tracking performance can be gained. So base on the work done in this paper, the telecontrol of excavator can be realized if communication system is developed; and automatic excavation can be achieved if the trajectory programming is completed.

Appendix:

Fig.8 The experimental robotic excavator based on
Sunward SWE85 excavator

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